

LECTURES
ON
MODERN CONVEX OPTIMIZATION

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Preface

Mathematical Programming deals with optimization programs of the form

$$\begin{aligned} & \text{minimize} && f(x) \\ & \text{subject to} && g_i(x) \leq 0, i = 1, \dots, m, \\ & && [x \subset \mathbf{R}^n] \end{aligned} \tag{P}$$

and includes the following general areas:

1. Modelling: methodologies for posing various applied problems as optimization programs;
2. Optimization Theory, focusing on existence, uniqueness and on characterization of optimal solutions to optimization programs;
3. Optimization Methods: development and analysis of computational algorithms for various classes of optimization programs;
4. Implementation, testing and application of modelling methodologies and computational algorithms.

Essentially, Mathematical Programming was born in 1948, when George Dantzig has invented *Linear Programming* – the class of optimization programs (P) with linear objective $f(\cdot)$ and constraints $g_i(\cdot)$. This breakthrough discovery included

- the *methodological* idea that a natural desire of a human being to look for the best possible decisions can be posed in the form of an optimization program (P) and thus subject to mathematical and computational treatment;
- the *theory* of LP programs, primarily the LP duality (this is in part due to the great mathematician John von Neumann);
- the first *computational method* for LP – the Simplex method, which over the years turned out to be an extremely powerful computational tool.

As it often happens with first-rate discoveries (and to some extent is characteristic for such discoveries), today the above ideas and constructions look quite traditional and simple. Well, the same is with the wheel.

In 50 plus years since its birth, Mathematical Programming was rapidly progressing along all outlined avenues, “in width” as well as “in depth”. We have no intention (and time) to trace the history of the subject decade by decade; instead, let us outline the major achievements in Optimization during the last 20 years or so, those which, we believe, allow to speak about *modern* optimization as opposed to the “classical” one as it existed circa 1980. The reader should be aware that the summary to follow is highly subjective and reflects the personal preferences of the authors. Thus, *in our opinion* the major achievements in Mathematical Programming during last 15-20 years can be outlined as follows:

♠ Realizing what are the generic optimization programs one can solve well (“efficiently solvable” programs) and when such a possibility is, mildly speaking, problematic (“computationally intractable” programs). At this point, we do not intend to explain what does it mean exactly that “a generic optimization program is efficiently solvable”; we will arrive at this issue further in the course. However, we intend to answer the question (right now, not well posed!) “what are generic optimization programs we can solve well”:

(!) As far as numerical processing of programs (P) is concerned, there exists a “solvable case” – the one of convex optimization programs, where the objective f and the constraints g_i are convex functions.

Under minimal additional “computability assumptions” (which are satisfied in basically all applications), a convex optimization program is “computationally tractable” – the computational effort required to solve the problem to a given accuracy “grows moderately” with the dimensions of the problem and the required number of accuracy digits.

In contrast to this, a general-type non-convex problems are too difficult for numerical solution – the computational effort required to solve such a problem by the best known so far numerical methods grows prohibitively fast with the dimensions of the problem and the number of accuracy digits, and there are serious theoretical reasons to guess that this is an intrinsic feature of non-convex problems rather than a drawback of the existing optimization techniques.

Just to give an example, consider a pair of optimization problems. The first is

$$\begin{aligned} \text{minimize} \quad & -\sum_{i=1}^n x_i \\ \text{subject to} \quad & x_i^2 - x_i = 0, \quad i = 1, \dots, n; \\ & x_i x_j = 0 \quad \forall (i, j) \in \Gamma, \end{aligned} \tag{A}$$

Γ being a given set of pairs (i, j) of indices i, j . This is a fundamental combinatorial problem of computing the stability number of a graph; the corresponding “covering story” is as follows:

Assume that we are given n letters which can be sent through a telecommunication channel, say, $n = 256$ usual bytes. When passing through the channel, an input letter can be corrupted by errors; as a result, two distinct input letters can produce the same output and thus not necessarily can be distinguished at the receiving end. Let Γ be the set of “dangerous pairs of letters” – pairs (i, j) of distinct letters i, j which can be converted by the channel into the same output. If we are interested in error-free transmission, we should restrict the set S of letters we actually use to be *independent* – such that no pair (i, j) with $i, j \in S$ belongs to Γ . And in order to utilize best of all the capacity of the channel, we are interested to use a maximal – with maximum possible number of letters – independent sub-alphabet. It turns out that the minus optimal value in (A) is exactly the cardinality of such a maximal independent sub-alphabet.

Our second problem is

$$\begin{aligned} \text{minimize} \quad & -2 \sum_{i=1}^k \sum_{j=1}^m c_{ij} x_{ij} + x_{00} \\ \text{subject to} \quad & \lambda_{\min} \left(\begin{bmatrix} x_1 & & & & \sum_{j=1}^m b_{pj} x_{1j} \\ & \ddots & & & \cdots \\ & & x_k & & \sum_{j=1}^m b_{pj} x_{kj} \\ \sum_{j=1}^m b_{pj} x_{1j} & \cdots & \sum_{j=1}^m b_{pj} x_{kj} & \sum_{j=1}^m b_{pj} x_{kj} & x_{00} \\ & & & & p = 1, \dots, N, \\ & & & & \sum_{i=1}^k x_i = 1, \end{bmatrix} \right) \geq 0, \end{aligned} \tag{B}$$

where $\lambda_{\min}(A)$ denotes the minimum eigenvalue of a symmetric matrix A . This problem is responsible for the design of a *truss* (a mechanical construction comprised of linked with each other thin elastic bars, like an electric mast, a bridge or the Eiffel Tower) capable to withstand best of all to k given loads.

When looking at the analytical forms of (A) and (B), it seems that the first problem is easier than the second: the constraints in (A) are simple explicit quadratic equations, while the constraints in (B) involve much more complicated functions of the design variables – the eigenvalues of certain matrices depending on the design vector. The truth, however, is that the first problem is, in a sense, “as difficult as an optimization problem can be”, and

the worst-case computational effort to solve this problem within absolute inaccuracy 0.5 by all known optimization methods is about 2^n operations; for $n = 256$ (just 256 design variables corresponding to the “alphabet of bytes”), the quantity $2^n \approx 10^{77}$, for all practical purposes, is the same as $+\infty$. In contrast to this, the second problem is quite “computationally tractable”. E.g., for $k = 6$ (6 loads of interest) and $m = 100$ (100 degrees of freedom of the construction) the problem has about 600 variables (twice the one of the “byte” version of (A)); however, it can be reliably solved within 6 accuracy digits in a couple of minutes. The dramatic difference in computational effort required to solve (A) and (B) finally comes from the fact that (A) is a non-convex optimization problem, while (B) is convex.

Note that realizing what is easy and what is difficult in Optimization is, aside of theoretical importance, extremely important methodologically. Indeed, mathematical models of real world situations in any case are incomplete and therefore are flexible to some extent. When you know in advance what you can process efficiently, you perhaps can use this flexibility to build a tractable (in our context – a convex) model. The “traditional” Optimization did not pay much attention to complexity and focused on easy-to-analyze *purely asymptotical* “rate of convergence” results. From this viewpoint, the most desirable property of f and g_i is smoothness (plus, perhaps, certain “nondegeneracy” at the optimal solution), and not their convexity; choosing between the above problems (A) and (B), a “traditional” optimizer would, perhaps, prefer the first of them. We suspect that a non-negligible part of “applied failures” of Mathematical Programming came from the traditional (we would say, heavily misleading) “order of preferences” in model-building. Surprisingly, some advanced users (primarily in Control) have realized the crucial role of convexity much earlier than some members of the Optimization community. Here is a real story. About 7 years ago, we were working on certain Convex Optimization method, and one of us sent an e-mail to people maintaining CUTE (a benchmark of test problems for constrained continuous optimization) requesting for the list of convex programs from their collection. The answer was: “We do not care which of our problems are convex, and this be a lesson for those developing Convex Optimization techniques.” In their opinion, the question is stupid; in our opinion, they are obsolete. Who is right, this we do not know...

♠ *Discovery of interior-point polynomial time methods for “well-structured” generic convex programs and throughout investigation of these programs.*

By itself, the “efficient solvability” of generic convex programs is a theoretical rather than a practical phenomenon. Indeed, assume that all we know about (P) is that the program is convex, its objective is called f , the constraints are called g_i and that we can compute f and g_i , along with their derivatives, at any given point at the cost of M arithmetic operations. In this case the computational effort for finding an ϵ -solution turns out to be at least $O(1)nM\ln(\frac{1}{\epsilon})$. Note that this is a *lower* complexity bound, and the best known so far *upper* bound is much worse: $O(1)n(n^3 + M)\ln(\frac{1}{\epsilon})$. Although the bounds grow “moderately” – polynomially – with the design dimension n of the program and the required number $\ln(\frac{1}{\epsilon})$ of accuracy digits, from the practical viewpoint the upper bound becomes prohibitively large already for n like 1000. This is in striking contrast with Linear Programming, where one can solve routinely problems with tens and hundreds of thousands of variables and constraints. The reasons for this huge difference come from the fact that

When solving an LP program, our a priory knowledge is far beyond the fact that the objective is called f , the constraints are called g_i , that they are convex and we can compute their values at derivatives at any given point. In LP, we know in advance what is the analytical structure of f and g_i , and we heavily exploit this knowledge when processing the problem. In fact, all successful LP methods never never compute the values and the derivatives of f and g_i – they do something completely different.

One of the most important recent developments in Optimization is realizing the simple fact that a *jump from linear f and g_i 's to “completely structureless” convex f and g_i 's is too long: in-between these two extremes, there are many interesting and important generic convex programs.* These “in-between” programs, although non-linear, still possess nice analytical structure, and one can use this structure to develop dedicated optimization methods, the methods which turn out to be incomparably more efficient than those exploiting solely the convexity of the program.

The aforementioned “dedicated methods” are *Interior Point polynomial time algorithms*, and the most important “well-structured” generic convex optimization programs are those of *Linear, Conic Quadratic and Semidefinite Programming*; the last two entities merely did not exist as established research subjects just 15 years ago. In our opinion, *the discovery of Interior Point methods and of non-linear “well-structured” generic convex programs, along with the subsequent progress in these novel research areas, is one of the most impressive achievements in Mathematical Programming*.

♠ We have outlined the most revolutionary, in our appreciation, changes in the *theoretical core of Mathematical Programming* in the last 15-20 years. During this period, we have witnessed perhaps less dramatic, but still quite important progress in the methodological and application-related areas as well. The major novelty here is certain shift from the traditional for Operations Research applications in Industrial Engineering (production planning, etc.) to applications in “genuine” Engineering. We believe it is completely fair to say that the theory and methods of Convex Optimization, especially those of Semidefinite Programming, have become a kind of new paradigm in Control and are becoming more and more frequently used in Mechanical Engineering, Design of Structures, Medical Imaging, etc.

The aim of the course is to outline some of the novel research areas which have arisen in Optimization during the past decade or so. We intend to focus solely on Convex Programming, specifically, on

- *Conic Programming*, with emphasis on the most important particular cases – those of Linear, Conic Quadratic and Semidefinite Programming (LP, CQP and SDP, respectively).

Here the focus will be on

- basic Duality Theory for conic programs;
- investigation of “expressive abilities” of CQP and SDP;
- overview of the theory of Interior Point polynomial time methods for LP, CQP and SDP.
- “Efficient (polynomial time) solvability” of generic convex programs.
- “Low cost” optimization methods for extremely large-scale optimization programs.

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of the subject expressed in this course. Needless to say, we are the only persons responsible for the drawbacks in what follows.

Aharon Ben-Tal, Arkadi Nemirovski,
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The Lecture Notes were “renovated” in Fall 2011, Summer 2013 and Summer 2015. The most important added material is the one in Sections 1.3 (applications of LP in Compressive Sensing and for synthesis of linear controllers in linear dynamic systems), 3.6 (Lagrangian approximation of chance constraints), and 5.3, 5.5 (advanced First Order methods for large-scale optimization). I have also added three “operational exercises” (Exercises 1.3.1, 1.3.2, 2.6.1, 5.3.1); in contrast to “regular exercises,” where the task usually is to prove something, an operational exercise requires creating software and numerical experimentation to implement a broadly outlined approach and as such offers much room for reader’s initiative. The operational exercises are typed in blue.

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Main Notational Conventions

Vectors and matrices. By default, all vectors are *column vectors*.

- The space of all n -dimensional vectors is denoted \mathbf{R}^n , the set of all $m \times n$ matrices is denoted $\mathbf{R}^{m \times n}$ or $\mathbf{M}^{m \times n}$, and the set of symmetric $n \times n$ matrices is denoted \mathbf{S}^n . By default, all vectors and matrices are real.
- Sometimes, “MATLAB notation” is used: a vector with coordinates x_1, \dots, x_n is written down as

$$x = [x_1; \dots; x_n]$$

(pay attention to semicolon “;”). For example, $\begin{bmatrix} 1 \\ 2 \\ 3 \end{bmatrix}$ is written as $[1; 2; 3]$.

More generally,

— if A_1, \dots, A_m are matrices with the same number of columns, we write $[A_1; \dots; A_m]$ to denote the matrix which is obtained when writing A_2 beneath A_1 , A_3 beneath A_2 , and so on.

— if A_1, \dots, A_m are matrices with the same number of rows, then $[A_1, \dots, A_m]$ stands for the matrix which is obtained when writing A_2 to the right of A_1 , A_3 to the right of A_2 , and so on.

Examples:

- $A_1 = \begin{bmatrix} 1 & 2 & 3 \\ 4 & 5 & 6 \end{bmatrix}$, $A_2 = \begin{bmatrix} 7 & 8 & 9 \end{bmatrix} \Rightarrow [A_1; A_2] = \begin{bmatrix} 1 & 2 & 3 \\ 4 & 5 & 6 \\ 7 & 8 & 9 \end{bmatrix}$
- $A_1 = \begin{bmatrix} 1 & 2 \\ 3 & 4 \end{bmatrix}$, $A_2 = \begin{bmatrix} 7 \\ 8 \end{bmatrix} \Rightarrow [A_1, A_2] = \begin{bmatrix} 1 & 2 & 7 \\ 4 & 5 & 8 \end{bmatrix}$
- $[1, 2, 3, 4] = [1; 2; 3; 4]^T$
- $[[1, 2; 3, 4], [5, 6; 7, 8]] = \left[\begin{bmatrix} 1 & 2 \\ 3 & 4 \end{bmatrix}, \begin{bmatrix} 5 & 6 \\ 7 & 8 \end{bmatrix} \right] = \begin{bmatrix} 1 & 2 & 5 & 6 \\ 3 & 4 & 7 & 8 \end{bmatrix}$
 $= [1, 2, 5, 6; 3, 4, 7, 8]$

O(1)’s. Below $O(1)$ ’s denote properly selected *positive absolute constants*. We write $f \leq O(1)g$, where f and g are nonnegative functions of some parameters, to express the fact that for properly selected positive absolute constant C the inequality $f \leq Cg$ holds true in the entire range of the parameters, and we write $f = O(1)g$ when both $f \leq O(1)g$ and $g \leq O(1)f$.

Lecture 1

From Linear to Conic Programming

1.1 Linear programming: basic notions

A *Linear Programming* (LP) program is an optimization program of the form

$$\min \left\{ c^T x \mid Ax \geq b \right\}, \quad (\text{LP})$$

where

- $x \in \mathbf{R}^n$ is the *design vector*
- $c \in \mathbf{R}^n$ is a given vector of coefficients of the *objective function* $c^T x$
- A is a given $m \times n$ *constraint matrix*, and $b \in \mathbf{R}^m$ is a given *right hand side of the constraints*.

(LP) is called

- *feasible*, if its *feasible set*

$$\mathcal{F} = \{x : Ax - b \geq 0\}$$

is nonempty; a point $x \in \mathcal{F}$ is called a *feasible solution* to (LP);

- *bounded below*, if it is either infeasible, or its objective $c^T x$ is bounded below on \mathcal{F} .

For a feasible bounded below problem (LP), the quantity

$$c^* \equiv \inf_{x:Ax-b \geq 0} c^T x$$

is called the *optimal value* of the problem. For an infeasible problem, we set $c_* = +\infty$, while for feasible unbounded below problem we set $c_* = -\infty$.

(LP) is called *solvable*, if it is feasible, bounded below and the optimal value is attained, i.e., there exists $x \in \mathcal{F}$ with $c^T x = c^*$. An x of this type is called an *optimal solution* to (LP).

A priori it is unclear whether a feasible and bounded below LP program is solvable: why should the infimum be achieved? It turns out, however, that a feasible and bounded below program (LP) always is solvable. This nice fact (we shall establish it later) is specific for LP. Indeed, a very simple *nonlinear* optimization program

$$\min \left\{ \frac{1}{x} \mid x \geq 1 \right\}$$

is feasible and bounded below, but it is not solvable.

1.2 Duality in Linear Programming

The most important and interesting feature of linear programming as a mathematical entity (i.e., aside of computations and applications) is the wonderful *LP duality theory* we are about to consider. We motivate this topic by first addressing the following question:

Given an LP program

$$c^* = \min_x \left\{ c^T x \mid Ax - b \geq 0 \right\}, \quad (\text{LP})$$

how to find a systematic way to bound from below its optimal value c^ ?*

Why this is an important question, and how the answer helps to deal with LP, this will be seen in the sequel. For the time being, let us just believe that the question is worthy of the effort.

A trivial answer to the posed question is: solve (LP) and look what is the optimal value. There is, however, a smarter and a much more instructive way to answer our question. Just to get an idea of this way, let us look at the following example:

$$\min \left\{ \begin{array}{l} x_1 + 2x_2 + \dots + 2001x_{2001} + 2002x_{2002} - 1 \geq 0, \\ x_1 + x_2 + \dots + x_{2002} \mid 2002x_1 + 2001x_2 + \dots + 2x_{2001} + x_{2002} - 100 \geq 0, \\ \dots \dots \dots \end{array} \right\}.$$

We claim that the optimal value in the problem is $\geq \frac{101}{2003}$. How could one certify this bound? This is immediate: add the first two constraints to get the inequality

$$2003(x_1 + x_2 + \dots + x_{1998} + x_{2002}) - 101 \geq 0,$$

and divide the resulting inequality by 2003. LP duality is nothing but a straightforward generalization of this simple trick.

1.2.1 Certificates for solvability and insolvability

Consider a (finite) system of scalar inequalities with n unknowns. To be as general as possible, we do not assume for the time being the inequalities to be linear, and we allow for both non-strict and strict inequalities in the system, as well as for equalities. Since an equality can be represented by a pair of non-strict inequalities, our system can always be written as

$$f_i(x) \Omega_i 0, \quad i = 1, \dots, m, \quad (\mathcal{S})$$

where every Ω_i is either the relation " $>$ " or the relation " \geq ".

The basic question about (\mathcal{S}) is

(?) Whether (\mathcal{S}) has a solution or not.

Knowing how to answer the question (?), we are able to answer many other questions. E.g., to verify whether a given real a is a lower bound on the optimal value c^* of (LP) is the same as to verify whether the system

$$\begin{cases} -c^T x + a > 0 \\ Ax - b \geq 0 \end{cases}$$

has no solutions.

The general question above is too difficult, and it makes sense to pass from it to a seemingly simpler one:

(??) How to certify that (\mathcal{S}) has, or does not have, a solution.

Imagine that you are very smart and know the correct answer to (?); how could you convince somebody that your answer is correct? What could be an “evident for everybody” certificate of the validity of your answer?

If your claim is that (\mathcal{S}) is solvable, a certificate could be just to point out a solution x^* to (\mathcal{S}) . Given this certificate, one can substitute x^* into the system and check whether x^* indeed is a solution.

Assume now that your claim is that (\mathcal{S}) has no solutions. What could be a “simple certificate” of this claim? How one could certify a negative statement? This is a highly nontrivial problem not just for mathematics; for example, in criminal law: how should someone accused in a murder prove his innocence? The “real life” answer to the question “how to certify a negative statement” is discouraging: such a statement normally *cannot* be certified (this is where the rule “a person is presumed innocent until proven guilty” comes from). In mathematics, however, the situation is different: in some cases there exist “simple certificates” of negative statements. E.g., in order to certify that (\mathcal{S}) has no solutions, it suffices to demonstrate that a consequence of (\mathcal{S}) is a contradictory inequality such as

$$-1 \geq 0.$$

For example, assume that $\lambda_i, i = 1, \dots, m$, are nonnegative weights. Combining inequalities from (\mathcal{S}) with these weights, we come to the inequality

$$\sum_{i=1}^m \lambda_i f_i(x) \Omega 0 \quad (\text{Cons}(\lambda))$$

where Ω is either ” $>$ ” (this is the case when the weight of at least one strict inequality from (\mathcal{S}) is positive), or ” \geq ” (otherwise). Since the resulting inequality, due to its origin, is a consequence of the system (\mathcal{S}) , i.e., it is satisfied by every solution to \mathcal{S}), it follows that if $(\text{Cons}(\lambda))$ has no solutions at all, we can be sure that (\mathcal{S}) has no solution. Whenever this is the case, we may treat the corresponding vector λ as a “simple certificate” of the fact that (\mathcal{S}) is infeasible.

Let us look what does the outlined approach mean when (\mathcal{S}) is comprised of *linear* inequalities:

$$(\mathcal{S}) : \quad \{a_i^T x \Omega_i b_i, i = 1, \dots, m\} \quad \left[\Omega_i = \begin{cases} " > " \\ " \geq " \end{cases} \right]$$

Here the “combined inequality” is linear as well:

$$(\text{Cons}(\lambda)) : \quad \left(\sum_{i=1}^m \lambda a_i \right)^T x \Omega \sum_{i=1}^m \lambda b_i$$

(Ω is ” $>$ ” whenever $\lambda_i > 0$ for at least one i with $\Omega_i = "$ $>$ ”, and Ω is ” \geq ” otherwise). Now, when can a *linear* inequality

$$d^T x \Omega e$$

be contradictory? Of course, it can happen only when $d = 0$. Whether in this case the inequality is contradictory, it depends on what is the relation Ω : if $\Omega = "$ $>$ ”, then the inequality is contradictory if and only if $e \geq 0$, and if $\Omega = "$ \geq ”, it is contradictory if and only if $e > 0$. We have established the following simple result:

Proposition 1.2.1 Consider a system of linear inequalities

$$(\mathcal{S}) : \quad \begin{cases} a_i^T x > b_i, & i = 1, \dots, m_s, \\ a_i^T x \geq b_i, & i = m_s + 1, \dots, m. \end{cases}$$

with n -dimensional vector of unknowns x . Let us associate with (\mathcal{S}) two systems of linear inequalities and equations with m -dimensional vector of unknowns λ :

$$\begin{aligned} \mathcal{T}_I : & \quad \begin{cases} (a) \quad \lambda \geq 0; \\ (b) \quad \sum_{i=1}^m \lambda_i a_i = 0; \\ (c_I) \quad \sum_{i=1}^m \lambda_i b_i \geq 0; \\ \hline (d_I) \quad \sum_{i=1}^{m_s} \lambda_i > 0. \end{cases} \\ \mathcal{T}_{II} : & \quad \begin{cases} (a) \quad \lambda \geq 0; \\ (b) \quad \sum_{i=1}^m \lambda_i a_i = 0; \\ \hline (c_{II}) \quad \sum_{i=1}^m \lambda_i b_i > 0. \end{cases} \end{aligned}$$

Assume that at least one of the systems \mathcal{T}_I , \mathcal{T}_{II} is solvable. Then the system (\mathcal{S}) is infeasible.

Proposition 1.2.1 says that in some cases it is easy to certify infeasibility of a linear system of inequalities: a “simple certificate” is a solution to another system of linear inequalities. Note, however, that the existence of a certificate of this latter type is to the moment only a *sufficient*, but not a *necessary*, condition for the infeasibility of (\mathcal{S}) . A fundamental result in the theory of linear inequalities is that the sufficient condition in question is in fact also necessary:

Theorem 1.2.1 [General Theorem on Alternative] In the notation from Proposition 1.2.1, system (\mathcal{S}) has no solutions if and only if either \mathcal{T}_I , or \mathcal{T}_{II} , or both these systems, are solvable.

There are numerous proofs of the Theorem on Alternative; in my taste, the most instructive one is to reduce the Theorem to its particular case – the *Homogeneous Farkas Lemma*:

[Homogeneous Farkas Lemma] A homogeneous nonstrict linear inequality

$$a^T x \leq 0$$

is a consequence of a system of homogeneous nonstrict linear inequalities

$$a_i^T x \leq 0, \quad i = 1, \dots, m$$

if and only if it can be obtained from the system by taking weighted sum with nonnegative weights:

$$\begin{aligned} (a) \quad a_i^T x \leq 0, \quad i = 1, \dots, m \Rightarrow a^T x \leq 0, \\ (b) \quad \exists \lambda_i \geq 0 : a = \sum_i \lambda_i a_i. \end{aligned} \quad (1.2.1)$$

The reduction of GTA to HFL is easy. As about the HFL, there are, essentially, two ways to prove the statement:

- The “quick and dirty” one based on separation arguments (see Section B.2.6 and/or Exercise B.14), which is as follows:

1. First, we demonstrate that if A is a nonempty closed convex set in \mathbf{R}^n and a is a point from $\mathbf{R}^n \setminus A$, then a can be *strongly separated* from A by a linear form: there exists $x \in \mathbf{R}^n$ such that

$$x^T a < \inf_{b \in A} x^T b. \quad (1.2.2)$$

To this end, it suffices to verify that

- (a) In A , there exists a point closest to a w.r.t. the standard Euclidean norm $\|b\|_2 = \sqrt{b^T b}$, i.e., that the optimization program

$$\min_{b \in A} \|a - b\|_2$$

has a solution b_* ;

- (b) Setting $x = b_* - a$, one ensures (1.2.2).

Both (a) and (b) are immediate.

2. Second, we demonstrate that *the set*

$$A = \{b : \exists \lambda \geq 0 : b = \sum_{i=1}^m \lambda_i a_i\}$$

– the cone spanned by the vectors a_1, \dots, a_m – is convex (which is immediate) and closed (the proof of this crucial fact also is not difficult).

3. Combining the above facts, we immediately see that

- either $a \in A$, i.e., (1.2.1.b) holds,
- or there exists x such that $x^T a < \inf_{\lambda \geq 0} x^T \sum_i \lambda_i a_i$.

The latter inf is finite if and only if $x^T a_i \geq 0$ for all i , and in this case the inf is 0, so that the “or” statement says exactly that there exists x with $a_i^T x \geq 0$, $a^T x < 0$, or, which is the same, that (1.2.1.a) does not hold.

Thus, among the statements (1.2.1.a) and the negation of (1.2.1.b) at least one (and, as it is immediately seen, at most one as well) always is valid, which is exactly the equivalence (1.2.1).

- “Advanced” proofs based purely on Linear Algebra facts (see Section B.2.5.A). The advantage of these purely Linear Algebra proofs is that they, in contrast to the outlined separation-based proof, do not use the completeness of \mathbf{R}^n as a metric space and thus work when we pass from systems with *real* coefficients and unknowns to systems with *rational* (or algebraic) coefficients. As a result, an advanced proof allows to establish the Theorem on Alternative for the case when the coefficients and unknowns in (\mathcal{S}) , \mathcal{T}_I , \mathcal{T}_{II} are restricted to belong to a given “real field” (e.g., are rational).

We formulate here explicitly two very useful principles following from the Theorem on Alternative:

A. A system of linear inequalities

$$a_i^T x \leq b_i, \quad i = 1, \dots, m$$

has no solutions if and only if one can combine the inequalities of the system in a linear fashion (i.e., multiplying the inequalities by nonnegative weights, adding the results and passing, if necessary, from an inequality $a^T x > b$ to the inequality $a^T x \geq b$) to get a contradictory inequality, namely, either the inequality $0^T x \geq 1$, or the inequality $0^T x > 0$.

B. A linear inequality

$$a_0^T x \leq b_0$$

is a consequence of a solvable system of linear inequalities

$$a_i^T x \leq b_i, \quad i = 1, \dots, m$$

if and only if it can be obtained by combining, in a linear fashion, the inequalities of the system and the trivial inequality $0 > -1$.

It should be stressed that the above principles are highly nontrivial and very deep. Consider, e.g., the following system of 4 linear inequalities with two variables u, v :

$$\begin{aligned} -1 &\leq u \leq 1 \\ -1 &\leq v \leq 1. \end{aligned}$$

From these inequalities it follows that

$$u^2 + v^2 \leq 2, \quad (!)$$

which in turn implies, by the Cauchy inequality, the linear inequality $u + v \leq 2$:

$$u + v = 1 \times u + 1 \times v \leq \sqrt{1^2 + 1^2} \sqrt{u^2 + v^2} \leq (\sqrt{2})^2 = 2. \quad (!!)$$

The concluding inequality is linear and is a consequence of the original system, but in the demonstration of this fact both steps (!) and (!!) are “highly nonlinear”. It is absolutely unclear a priori why the same consequence can, as it is stated by Principle **A**, be derived from the system in a linear manner as well [of course it can – it suffices just to add two inequalities $u \leq 1$ and $v \leq 1$].

Note that the Theorem on Alternative and its corollaries **A** and **B** heavily exploit the fact that we are speaking about *linear* inequalities. E.g., consider the following 2 quadratic and 2 linear inequalities with two variables:

$$\begin{array}{lll} (a) & u^2 & \geq 1; \\ (b) & v^2 & \geq 1; \\ (c) & u & \geq 0; \\ (d) & v & \geq 0; \end{array}$$

along with the quadratic inequality

$$(e) \quad uv \geq 1.$$

The inequality (e) is clearly a consequence of (a) – (d). However, if we extend the system of inequalities (a) – (b) by all “trivial” (i.e., identically true) linear and quadratic inequalities with 2 variables, like $0 > -1$, $u^2 + v^2 \geq 0$, $u^2 + 2uv + v^2 \geq 0$, $u^2 - uv + v^2 \geq 0$, etc., and ask whether (e) can be derived in a *linear* fashion from the inequalities of the extended system, the answer will be negative. Thus, Principle **A** fails to be true already for quadratic inequalities (which is a great sorrow – otherwise there were no difficult problems at all!)

We are about to use the Theorem on Alternative to obtain the basic results of the LP duality theory.

1.2.2 Dual to an LP program: the origin

As already mentioned, the motivation for constructing the problem dual to an LP program

$$c^* = \min_x \left\{ c^T x \mid Ax - b \geq 0 \right\} \quad \left[A = \begin{bmatrix} a_1^T \\ a_2^T \\ \vdots \\ a_m^T \end{bmatrix} \in \mathbf{R}^{m \times n} \right] \quad (\text{LP})$$

is the desire to generate, in a systematic way, lower bounds on the optimal value c^* of (LP). An evident way to bound from below a given function $f(x)$ in the domain given by system of inequalities

$$g_i(x) \geq b_i, i = 1, \dots, m, \quad (1.2.3)$$

is offered by what is called the *Lagrange duality* and is as follows:

Lagrange Duality:

- Let us look at all inequalities which can be obtained from (1.2.3) by linear aggregation, i.e., at the inequalities of the form

$$\sum_i y_i g_i(x) \geq \sum_i y_i b_i \quad (1.2.4)$$

with the “aggregation weights” $y_i \geq 0$. Note that the inequality (1.2.4), due to its origin, is valid on the entire set X of solutions of (1.2.3).

- Depending on the choice of aggregation weights, it may happen that the left hand side in (1.2.4) is $\leq f(x)$ for all $x \in \mathbf{R}^n$. Whenever it is the case, the right hand side $\sum_i y_i b_i$ of (1.2.4) is a lower bound on f in X .

Indeed, on X the quantity $\sum_i y_i b_i$ is a lower bound on $y_i g_i(x)$, and for y in question the latter function of x is everywhere $\leq f(x)$.

It follows that

- The optimal value in the problem

$$\max_y \left\{ \sum_i y_i b_i : \begin{array}{l} y \geq 0, \\ \sum_i y_i g_i(x) \leq f(x) \quad \forall x \in \mathbf{R}^n \end{array} \right. \begin{array}{l} (a) \\ (b) \end{array} \left. \right\} \quad (1.2.5)$$

is a lower bound on the values of f on the set of solutions to the system (1.2.3).

Let us look what happens with the Lagrange duality when f and g_i are homogeneous linear functions: $f = c^T x$, $g_i(x) = a_i^T x$. In this case, the requirement (1.2.5.b) merely says that $c = \sum_i y_i a_i$ (or, which is the same, $A^T y = c$ due to the origin of A). Thus, problem (1.2.5) becomes the Linear Programming problem

$$\max_y \left\{ b^T y : A^T y = c, y \geq 0 \right\}, \quad (\text{LP}^*)$$

which is nothing but the LP dual of (LP).

By the construction of the dual problem,

[Weak Duality] The optimal value in (LP^*) is less than or equal to the optimal value in (LP).

In fact, the “less than or equal to” in the latter statement is “equal”, provided that the optimal value c^* in (LP) is a number (i.e., (LP) is feasible and below bounded). To see that this indeed is the case, note that a real a is a lower bound on c^* if and only if $c^T x \geq a$ whenever $Ax \geq b$, or, which is the same, if and only if the system of linear inequalities

$$(\mathcal{S}_a) : \quad -c^T x > -a, Ax \geq b$$

has no solution. We know by the Theorem on Alternative that the latter fact means that some other system of linear equalities (more exactly, at least one of a certain pair of systems) does have a solution. More precisely,

(*) (\mathcal{S}_a) has no solutions if and only if at least one of the following two systems with $m + 1$ unknowns:

$$\mathcal{T}_I : \left\{ \begin{array}{l} (a) \quad \lambda = (\lambda_0, \lambda_1, \dots, \lambda_m) \geq 0; \\ (b) \quad -\lambda_0 c + \sum_{i=1}^m \lambda_i a_i = 0; \\ (c_I) \quad -\lambda_0 a + \sum_{i=1}^m \lambda_i b_i \geq 0; \\ (d_I) \quad \lambda_0 > 0, \end{array} \right.$$

or

$$\mathcal{T}_{II} : \left\{ \begin{array}{l} (a) \quad \lambda = (\lambda_0, \lambda_1, \dots, \lambda_m) \geq 0; \\ (b) \quad -\lambda_0 c - \sum_{i=1}^m \lambda_i a_i = 0; \\ (c_{II}) \quad -\lambda_0 a - \sum_{i=1}^m \lambda_i b_i > 0 \end{array} \right.$$

– has a solution.

Now assume that (LP) is feasible. We claim that under this assumption (\mathcal{S}_a) has no solutions if and only if \mathcal{T}_I has a solution.

The implication " \mathcal{T}_I has a solution $\Rightarrow (\mathcal{S}_a)$ has no solution" is readily given by the above remarks. To verify the inverse implication, assume that (\mathcal{S}_a) has no solutions and the system $Ax \leq b$ has a solution, and let us prove that then \mathcal{T}_I has a solution. If \mathcal{T}_I has no solution, then by (*) \mathcal{T}_{II} has a solution and, moreover, $\lambda_0 = 0$ for (every) solution to \mathcal{T}_{II} (since a solution to the latter system with $\lambda_0 > 0$ solves \mathcal{T}_I as well). But the fact that \mathcal{T}_{II} has a solution λ with $\lambda_0 = 0$ is independent of the values of a and c ; if this fact would take place, it would mean, by the same Theorem on Alternative, that, e.g., the following instance of (\mathcal{S}_a) :

$$0^T x \geq -1, Ax \geq b$$

has no solutions. The latter means that the system $Ax \geq b$ has no solutions – a contradiction with the assumption that (LP) is feasible. ■

Now, if \mathcal{T}_I has a solution, this system has a solution with $\lambda_0 = 1$ as well (to see this, pass from a solution λ to the one λ/λ_0 ; this construction is well-defined, since $\lambda_0 > 0$ for every solution to \mathcal{T}_I). Now, an $(m + 1)$ -dimensional vector $\lambda = (1, y)$ is a solution to \mathcal{T}_I if and only if the m -dimensional vector y solves the system of linear inequalities and equations

$$\begin{aligned} y &\geq 0; \\ A^T y \equiv \sum_{i=1}^m y_i a_i &= c; \\ b^T y &\geq a \end{aligned} \tag{D}$$

Summarizing our observations, we come to the following result.

Proposition 1.2.2 Assume that system (D) associated with the LP program (LP) has a solution (y, a) . Then a is a lower bound on the optimal value in (LP). Vice versa, if (LP) is feasible and a is a lower bound on the optimal value of (LP), then a can be extended by a properly chosen m -dimensional vector y to a solution to (D).

We see that the entity responsible for lower bounds on the optimal value of (LP) is the system (D): every solution to the latter system induces a bound of this type, and *in the case when (LP) is feasible*, all lower bounds can be obtained from solutions to (D). Now note that if (y, a) is a solution to (D), then the pair $(y, b^T y)$ also is a solution to the same system, and the lower bound $b^T y$ on c^* is not worse than the lower bound a . Thus, as far as lower bounds on c^* are concerned, we lose nothing by restricting ourselves to the solutions (y, a) of (D) with $a = b^T y$; the best lower bound on c^* given by (D) is therefore the optimal value of the problem $\max_y \{ b^T y \mid A^T y = c, y \geq 0 \}$, which is nothing but the dual to (LP) problem (LP*). Note that (LP*) is also a Linear Programming program.

All we know about the dual problem to the moment is the following:

Proposition 1.2.3 *Whenever y is a feasible solution to (LP*), the corresponding value of the dual objective $b^T y$ is a lower bound on the optimal value c^* in (LP). If (LP) is feasible, then for every $a \leq c^*$ there exists a feasible solution y of (LP*) with $b^T y \geq a$.*

1.2.3 The LP Duality Theorem

Proposition 1.2.3 is in fact equivalent to the following

Theorem 1.2.2 [Duality Theorem in Linear Programming] *Consider a linear programming program*

$$\min_x \{ c^T x \mid Ax \geq b \} \quad (\text{LP})$$

along with its dual

$$\max_y \{ b^T y \mid A^T y = c, y \geq 0 \} \quad (\text{LP}^*)$$

Then

- 1) The duality is symmetric: the problem dual to dual is equivalent to the primal;
- 2) The value of the dual objective at every dual feasible solution is \leq the value of the primal objective at every primal feasible solution
- 3) The following 5 properties are equivalent to each other:
 - (i) The primal is feasible and bounded below.
 - (ii) The dual is feasible and bounded above.
 - (iii) The primal is solvable.
 - (iv) The dual is solvable.
 - (v) Both primal and dual are feasible.

Whenever (i) \equiv (ii) \equiv (iii) \equiv (iv) \equiv (v) is the case, the optimal values of the primal and the dual problems are equal to each other.

Proof. 1) is quite straightforward: writing the dual problem (LP*) in our standard form, we get

$$\min_y \left\{ -b^T y \mid \begin{bmatrix} I_m \\ A^T \\ -A^T \end{bmatrix} y - \begin{pmatrix} 0 \\ -c \\ c \end{pmatrix} \geq 0 \right\},$$

where I_m is the m -dimensional unit matrix. Applying the duality transformation to the latter problem, we come to the problem

$$\max_{\xi, \eta, \zeta} \left\{ 0^T \xi + c^T \eta + (-c)^T \zeta \mid \begin{array}{l} \xi \geq 0 \\ \eta \geq 0 \\ \zeta \geq 0 \\ \xi - A\eta + A\zeta = -b \end{array} \right\},$$

which is clearly equivalent to (LP) (set $x = \eta - \zeta$).

2) is readily given by Proposition 1.2.3.

3):

(i) \Rightarrow (iv): If the primal is feasible and bounded below, its optimal value c^* (which of course is a lower bound on itself) can, by Proposition 1.2.3, be (non-strictly) majorized by a quantity $b^T y^*$, where y^* is a feasible solution to (LP*). In the situation in question, of course, $b^T y^* = c^*$ (by already proved item 2)); on the other hand, in view of the same Proposition 1.2.3, the optimal value in the dual is $\leq c^*$. We conclude that the optimal value in the dual is attained and is equal to the optimal value in the primal.

(iv) \Rightarrow (ii): evident;

(ii) \Rightarrow (iii): This implication, in view of the primal-dual symmetry, follows from the implication (i) \Rightarrow (iv).

(iii) \Rightarrow (i): evident.

We have seen that (i) \equiv (ii) \equiv (iii) \equiv (iv) and that the first (and consequently each) of these 4 equivalent properties implies that the optimal value in the primal problem is equal to the optimal value in the dual one. All which remains is to prove the equivalence between (i)–(iv), on one hand, and (v), on the other hand. This is immediate: (i)–(iv), of course, imply (v); vice versa, in the case of (v) the primal is not only feasible, but also bounded below (this is an immediate consequence of the feasibility of the dual problem, see 2)), and (i) follows. ■

An immediate corollary of the LP Duality Theorem is the following *necessary and sufficient* optimality condition in LP:

Theorem 1.2.3 [Necessary and sufficient optimality conditions in linear programming] *Consider an LP program (LP) along with its dual (LP*). A pair (x, y) of primal and dual feasible solutions is comprised of optimal solutions to the respective problems if and only if*

$$y_i [Ax - b]_i = 0, \quad i = 1, \dots, m, \quad [\text{complementary slackness}]$$

likewise as if and only if

$$c^T x - b^T y = 0 \quad [\text{zero duality gap}]$$

Indeed, the “zero duality gap” optimality condition is an immediate consequence of the fact that the value of primal objective at every primal feasible solution is \geq the value of the dual objective at every dual feasible solution, while the optimal values in the primal and the dual are equal to each other, see Theorem 1.2.2. The equivalence between the “zero duality gap” and the “complementary slackness” optimality conditions is given by the following

computation: whenever x is primal feasible and y is dual feasible, the products $y_i[Ax - b]_i$, $i = 1, \dots, m$, are nonnegative, while the sum of these products is precisely the duality gap:

$$y^T[Ax - b] = (A^T y)^T x - b^T y = c^T x - b^T y.$$

Thus, the duality gap can vanish at a primal-dual feasible pair (x, y) if and only if all products $y_i[Ax - b]_i$ for this pair are zeros.

1.3 Selected Engineering Applications of LP

Linear Programming possesses enormously wide spectrum of applications. Most of them, or at least the vast majority of applications presented in textbooks, have to do with Decision Making. Here we present an instructive sample of applications of LP in Engineering. The “common denominator” of what follows (except for the topic on Support Vector Machines, where we just tell stories) can be summarized as “LP Duality at work.”

1.3.1 Sparsity-oriented Signal Processing and ℓ_1 minimization

¹ Let us start with *Compressed Sensing* which addresses the problem as follows: “in the nature” there exists a *signal* represented by an n -dimensional vector x . We observe (perhaps, in the presence of observation noise) the image of x under linear transformation $x \mapsto Ax$, where A is a given $m \times n$ *sensing matrix*; thus, our observation is

$$y = Ax + \eta \in \mathbf{R}^m \tag{1.3.1}$$

where η is observation noise. Our goal is to recover x from the observed y . The outlined problem is responsible for an extremely wide variety of applications and, depending on a particular application, is studied in different “regimes.” For example, in the traditional Statistics x is interpreted not as a signal, but as the vector of parameters of a “black box” which, given on input a vector $a \in \mathbf{R}^n$, produces output $a^T x$. Given a collection a^1, \dots, a^m of n -dimensional inputs to the black box and the corresponding outputs (perhaps corrupted by noise) $y_i = [a^i]^T x + \eta_i$, we want to recover the vector of parameters x ; this is called *linear regression* problem,. In order to represent this problem in the form of (1.3.1) one should make the row vectors $[a^i]^T$ the rows of an $m \times n$ matrix, thus getting matrix A , and to set $y = [y_1; \dots; y_m]$, $\eta = [\eta_1; \dots; \eta_m]$. The typical regime here is $m \gg n$ - the number of observations is much larger than the number of parameters to be recovered, and the challenge is to use this “observation redundancy” in order to get rid, to the best extent possible, of the observation noise. In Compressed Sensing the situation is opposite: the regime of interest is $m \ll n$. At the first glance, this regime seems to be completely hopeless: even with no noise ($\eta = 0$), we need to recover a solution x to an *underdetermined* system of linear equations $y = Ax$. When the number of variables is greater than the number of observations, the solution to the system either does not exist, or is not unique, and in both cases our goal seems to be unreachable. This indeed is so, unless we have at our disposal some additional information on x . In Compressed Sensing, this additional information is that x is *s-sparse* — has at most a given number s of nonzero entries. Note that in many applications we indeed can be sure that the true signal x is sparse. Consider, e.g., the following story about signal detection:

¹For related reading, see, e.g., [16] and references therein.

There are n locations where signal transmitters could be placed, and m locations with the receivers. The contribution of a signal of unit magnitude originating in location j to the signal measured by receiver i is a known quantity a_{ij} , and signals originating in different locations merely sum up in the receivers; thus, if x is the n -dimensional vector with entries x_j representing the magnitudes of signals transmitted in locations $j = 1, 2, \dots, n$, then the m -dimensional vector y of (noiseless) measurements of the m receivers is $y = Ax$, $A \in \mathbf{R}^{m \times n}$. Given this vector, we intend to recover y .

Now, if the receivers are hydrophones registering noises emitted by submarines in certain part of Atlantic, tentative positions of submarines being discretized with resolution 500 m, the dimension of the vector x (the number of points in the discretization grid) will be in the range of tens of thousands, if not tens of millions. At the same time, the total number of submarines (i.e., nonzero entries in x) can be safely upper-bounded by 50, if not by 20.

1.3.1.1 Sparse recovery from deficient observations

Sparsity changes dramatically our possibilities to recover high-dimensional signals from their low-dimensional linear images: given in advance that x has at most $s \ll m$ nonzero entries, the possibility of exact recovery of x at least from noiseless observations y becomes quite natural. Indeed, let us try to recover x by the following “brute force” search: we inspect, one by one, all subsets I of the index set $\{1, \dots, n\}$ — first the empty set, then n singletons $\{1\}, \dots, \{n\}$, then $\frac{n(n-1)}{2}$ 2-element subsets, etc., and each time try to solve the system of linear equations

$$y = Ax, \quad x_j = 0 \text{ when } j \notin I;$$

when arriving for the first time at a solvable system, we terminate and claim that its solution is the true vector x . It is clear that we will terminate before all sets I of cardinality $\leq s$ are inspected. It is also easy to show (do it!) that if every $2s$ distinct columns in A are linearly independent (when $m \geq 2s$, this indeed is the case for a matrix A in a “general position”²), then the procedure is correct — it indeed recovers the true vector x .

A bad news is that the outlined procedure becomes completely impractical already for “small” values of s and n because of the astronomically large number of linear systems we need to process³. A partial remedy is as follows. The outlined approach is, essentially, a particular way to solve the optimization problem

$$\min\{\text{nnz}(x) : Ax = y\}, \tag{*}$$

²Here and in the sequel, the words “in general position” mean the following. We consider a family of objects, with a particular object — an instance of the family — identified by a vector of real parameters (you may think about the family of $n \times n$ square matrices; the vector of parameters in this case is the matrix itself). We say that an instance of the family possesses certain property *in general position*, if the set of values of the parameter vector for which the associated instance does *not* possess the property is of measure 0. Equivalently: randomly perturbing the parameter vector of an instance, the perturbation being uniformly distributed in a (whatever small) box, we with probability 1 get an instance possessing the property in question. E.g., a square matrix “in general position” is nonsingular.

³When $s = 5$ and $n = 100$, this number is $\approx 7.53e7$ — much, but perhaps doable. When $n = 200$ and $s = 20$, the number of systems to be processed jumps to $\approx 1.61e27$, which is by many orders of magnitude beyond our “computational grasp”; we would be unable to carry out that many computations even if the fate of the mankind were dependent on them. And from the perspective of Compressed Sensing, $n = 200$ still is a completely toy size, by 3-4 orders of magnitude less than we would like to handle.

where $\text{nnz}(x)$ is the number of nonzero entries in a vector x . At the present level of our knowledge, this problem looks completely intractable (in fact, we do not know algorithms solving the problem essentially faster than the brute force search), and there are strong reasons, to be addressed later in our course, to believe that it indeed is intractable. Well, if we do not know how to minimize under linear constraints the “bad” objective $\text{nnz}(x)$, let us “approximate” this objective with one which we do know how to minimize. The true objective is separable: $\text{nnz}(x) = \sum_{i=1}^n \xi(x_j)$, where $\xi(s)$ is the function on the axis equal to 0 at the origin and equal to 1 otherwise. As a matter of fact, the separable functions which we do know how to minimize under linear constraints are sums of convex functions of x_1, \dots, x_n ⁴. The most natural candidate to the role of convex approximation of $\xi(s)$ is $|s|$; with this approximation, (*) converts into the ℓ_1 -minimization problem

$$\min_x \left\{ \|x\|_1 := \sum_{i=1}^n |x_i| : Ax = y \right\}, \quad (1.3.2)$$

which is equivalent to the LP program

$$\min_{x,w} \left\{ \sum_{i=1}^n w_i : Ax = y, -w_i \leq x_i \leq w_i, 1 \leq i \leq n \right\}.$$

For the time being, we were focusing on the (unrealistic!) case of noiseless observations $\eta = 0$. A realistic model is that $\eta \neq 0$. How to proceed in this case, depends on what we know on η . In the simplest case of “unknown but small noise” one assumes that, say, the Euclidean norm $\|\cdot\|_2$ of η is upper-bounded by a given “noise level” δ : $\|\eta\|_2 \leq \delta$. In this case, the ℓ_1 recovery usually takes the form

$$\hat{x} = \operatorname{Argmin}_w \{ \|w\|_1 : \|Aw - y\|_2 \leq \delta \} \quad (1.3.3)$$

Now we cannot hope that our recovery \hat{x} will be exactly equal to the true s -sparse signal x , but perhaps may hope that \hat{x} is close to x when δ is small.

Note that (1.3.3) is not an LP program anymore⁵, but still is a nice convex optimization program which can be solved to high accuracy even for reasonable large m, n .

1.3.1.2 s -goodness and nullspace property

Let us say that a sensing matrix A is s -good, if in the noiseless case ℓ_1 minimization (1.3.2) recovers correctly all s -sparse signals x . It is easy to say when this is the case: the necessary and sufficient condition for A to be s -good is the following *nullspace property*:

$$\forall (z \in \mathbf{R}^n : Az = 0, z \neq 0, I \subset \{1, \dots, n\}, \text{Card}(I) \leq s) : \sum_{i \in I} |z_i| < \frac{1}{2} \|z\|_1. \quad (1.3.4)$$

In other words, for every nonzero vector $z \in \text{Ker } A$, the sum $\|z\|_{s,1}$ of the s largest magnitudes of entries in z should be strictly less than half of the sum of magnitudes of all entries.

⁴A real-valued function $f(s)$ on the real axis is called *convex*, if its graph, between every pair of its points, is below the chord linking these points, or, equivalently, if $f(x + \lambda(y - x)) \leq f(x) + \lambda(f(y) - f(x))$ for every $x, y \in \mathbf{R}$ and every $\lambda \in [0, 1]$. For example, maxima of (finitely many) affine functions $a_i s + b_i$ on the axis are convex. For more detailed treatment of convexity of functions, see Appendix C.

⁵To get an LP, we should replace the Euclidean norm $\|Aw - y\|_2$ of the residual with, say, the uniform norm $\|Aw - y\|_\infty$, which makes perfect sense when we start with coordinate-wise bounds on observation errors, which indeed is the case in some applications.

The necessity and sufficiency of the nullspace property for s -goodness of A can be derived “from scratch” — from the fact that s -goodness means that every s -sparse signal x should be the unique optimal solution to the associated LP $\min_w \{ \|w\|_1 : Aw = Ax\}$ combined with the LP optimality conditions. Another option, which we prefer to use here, is to guess the condition and then to prove that its indeed is necessary and sufficient for s -goodness of A . The necessity is evident: if the nullspace property does *not* take place, then there exists $0 \neq z \in \text{Ker}A$ and s -element subset I of the index set $\{1, \dots, n\}$ such that if J is the complement of I in $\{1, \dots, n\}$, then the vector z_I obtained from z by zeroing out all entries with indexes not in I along with the vector z_J obtained from z by zeroing out all entries with indexes not in J satisfy the relation $\|z_I\|_1 \geq \frac{1}{2}\|z\|_1 = \frac{1}{2}(\|z_I\|_1 + \|z_J\|_1)$, that is,

$$\|z_I\|_1 \geq \|z_J\|_1.$$

Since $Az = 0$, we have $Az_I = A[-z_J]$, and we conclude that the s -sparse vector z_I is *not* the unique optimal solution to the LP $\min_w \{ \|w\|_1 : Aw = Az_I\}$, since $-z_J$ is feasible solution to the program with the value of the objective at least as good as the one at z_J , on one hand, and the solution $-z_J$ is different from z_I (since otherwise we should have $z_I = z_J = 0$, whence $z = 0$, which is not the case) on the other hand.

To prove that the nullspace property is *sufficient* for A to be s -good is equally easy: indeed, assume that this property does take place, and let x be s -sparse signal, so that the indexes of nonzero entries in x are contained in an s -element subset I of $\{1, \dots, n\}$, and let us prove that if \hat{x} is an optimal solution to the LP (1.3.2), then $\hat{x} = X$. Indeed, denoting by J the complement of I setting $z = \hat{x} - x$ and assuming that $z \neq 0$, we have $Az = 0$. Further, in the same notation as above we have

$$\|x_I\|_1 - \|\hat{x}_I\|_1 \leq \|z_I\|_1 < \|z\|_J \leq \|\hat{x}_J\| - \|x_J\|_1$$

(the first and the third inequality are due to the Triangle inequality, and the second — due to the nullspace property), whence $\|x\|_1 = \|x_I\|_1 + \|x_J\|_1 < \|\hat{x}_I\|_1 + \|\hat{x}_J\| = \|\hat{x}\|_1$, which contradicts the origin of \hat{x} .

1.3.1.3 From nullspace property to error bounds for imperfect ℓ_1 recovery

The nullspace property establishes necessary and sufficient condition for the validity of ℓ_1 recovery in the noiseless case, whatever be the s -sparse true signal. We are about to show that after appropriate quantification, this property implies meaningful error bounds in the case of *imperfect* recovery (presence of observation noise, near-, but not exact, s -sparsity of the true signal, approximate minimization in (1.3.3)).

The aforementioned “proper quantification” of the nullspace property is suggested by the LP duality theory and is as follows. Let V_s be the set of all vectors $v \in \mathbf{R}^n$ with at most s nonzero entries, equal ± 1 each. Observing that the sum $\|z\|_{s,1}$ of the s largest magnitudes of entries in a vector z is nothing that $\max_{v \in V_s} v^T z$, the nullspace property says that the optimal value in the LP program

$$\gamma(v) = \max_z \{ v^T z : Az = 0, \|z\|_1 \leq 1 \} \quad (P_v)$$

is $< 1/2$ whenever $v \in V_s$ (why?). Applying the LP Duality Theorem, we get, after straightforward simplifications of the dual, that

$$\gamma(v) = \min_h \|A^T h - v\|_\infty.$$

Denoting by h_v an optimal solution to the right hand side LP, let us set

$$\gamma := \gamma_s(A) = \max_{v \in V_s} \gamma(v), \quad \beta_s(A) = \max_{v \in V_s} \|h_v\|_2.$$

Observe that the maxima in question are well defined reals, since V_s is a finite set, and that the nullspace property is nothing but the relation

$$\gamma_s(A) < 1/2. \quad (1.3.5)$$

Observe also that we have the following relation:

$$\forall z \in \mathbf{R}^n : \|z\|_{s,1} \leq \beta_s(A)\|Az\|_2 + \gamma_s(A)\|z\|_1. \quad (1.3.6)$$

Indeed, for $v \in V_s$ and $z \in \mathbf{R}^n$ we have

$$\begin{aligned} v^T z &= [v - A^T h_v]^T z + [A^T h_v]^T z \leq \|v - A^T h_v\|_\infty \|z\|_1 + h_v^T A z \\ &\leq \gamma(v)\|z\|_1 + \|h_v\|_2 \|Az\|_2 \leq \gamma_s(A)\|z\|_1 + \beta_s(A)\|Az\|_2. \end{aligned}$$

Since $\|z\|_{s,1} = \max_{v \in V_s} v^T z$, the resulting inequality implies (1.3.6).

Now consider imperfect ℓ_1 recovery $x \mapsto y \mapsto \hat{x}$, where

1. $x \in \mathbf{R}^n$ can be approximated within some accuracy ρ , measured in the ℓ_1 norm, by an s -sparse signal, or, which is the same,

$$\|x - x^s\|_1 \leq \rho$$

where x^s is the best s -sparse approximation of x (to get this approximation, one zeros out all but the s largest in magnitude entries in x , the ties, if any, being resolved arbitrarily);

2. y is a noisy observation of x :

$$y = Ax + \eta, \quad \|\eta\|_2 \leq \delta;$$

3. \hat{x} is a μ -suboptimal and ϵ -feasible solution to (1.3.3), specifically,

$$\|\hat{x}\|_1 \leq \mu + \min_w \{\|w\|_1 : \|Aw - y\|_2 \leq \delta\} \quad \& \quad \|A\hat{x} - y\|_2 \leq \epsilon.$$

Theorem 1.3.1 *Let A, s be given, and let the relation*

$$\forall z : \|z\|_{s,1} \leq \beta\|Az\|_2 + \gamma\|z\|_1 \quad (1.3.7)$$

holds true with some parameters $\gamma < 1/2$ and $\beta < \infty$ (as definitely is the case when A is s -good, $\gamma = \gamma_s(A)$ and $\beta = \beta_s(A)$). Then for the outlined imperfect ℓ_1 recovery the following error bound holds true:

$$\|\hat{x} - x\|_1 \leq \frac{2\beta(\delta + \epsilon) + \mu + 2\rho}{1 - 2\gamma}, \quad (1.3.8)$$

i.e., the recovery error is of order of the maximum of the “imperfections” mentioned in 1) — 3).

Proof. Let I be the set of indexes of the s largest in magnitude entries in x , J be the complement of I , and $z = \hat{x} - x$. Observing that x is feasible for (1.3.3), we have $\min_w \{\|w\|_1 : \|Aw - y\|_2 \leq \delta\} \leq \|x\|_1$, whence

$$\|\hat{x}\|_1 \leq \mu + \|x\|_1,$$

or, in the same notation as above,

$$\underbrace{\|x_I\|_1 - \|\hat{x}_I\|_1}_{\leq \|z_I\|_1} \geq \underbrace{\|\hat{x}_J\|_1 - \|x_J\|_1}_{\geq \|z_J\|_1 - 2\|x_J\|_1} - \mu$$

whence

$$\|z_J\|_1 \leq \mu + \|z_I\|_1 + 2\|x_J\|_1,$$

so that

$$\|z\|_1 \leq \mu + 2\|z_I\|_1 + 2\|x_J\|_1. \quad (a)$$

We further have

$$\|z_I\|_1 \leq \beta\|Az\|_2 + \gamma\|z\|_1,$$

which combines with (a) to imply that

$$\|z_I\|_1 \leq \beta\|Az\|_2 + \gamma[\mu + 2\|z_I\|_1 + 2\|x_J\|_1],$$

whence, in view of $\gamma < 1/2$ and due to $\|x_J\|_1 = \rho$,

$$\|z_I\|_1 \leq \frac{1}{1-2\gamma} [\beta\|Az\|_2 + \gamma[\mu + 2\rho]].$$

Combining this bound with (a), we get

$$\|z\|_1 \leq \mu + 2\rho + \frac{2}{1-2\gamma} [\beta\|Az\|_2 + \gamma[\mu + 2\rho]].$$

Recalling that $z = \hat{x} - x$ and that therefore $\|Az\|_2 \leq \|Ax - y\|_2 + \|A\hat{x} - y\|_2 \leq \delta + \epsilon$, we finally get

$$\|\hat{x} - x\|_1 \leq \mu + 2\rho + \frac{2}{1-2\gamma} [\beta[\delta + \epsilon] + \gamma[\mu + 2\rho]]. \quad \blacksquare$$

1.3.1.4 Compressed Sensing: Limits of performance

The Compressed Sensing theory demonstrates that

- For given m, n with $m \ll n$ (say, $m/n \leq 1/2$), there exist $m \times n$ sensing matrices which are s -good for the values of s “nearly as large as m ,” specifically, for $s \leq O(1) \frac{m}{\ln(n/m)}$ ⁶ Moreover, there are natural families of matrices where this level of goodness “is a rule.” E.g., when drawing an $m \times n$ matrix at random from the Gaussian or the ± 1 distributions (i.e., filling the matrix with independent realizations of a random variable which is either

⁶From now on, $O(1)$'s denote positive absolute constants – appropriately chosen numbers like 0.5, or 1, or perhaps 100,000. We could, in principle, replace all $O(1)$'s by specific numbers; following the standard mathematical practice, we do not do it, partly from laziness, partly because the particular values of these numbers in our context are irrelevant.

Gaussian (zero mean, variance $1/m$), or takes values $\pm 1/\sqrt{m}$ with probabilities 0.5⁷, the result will be s -good, for the outlined value of s , with probability approaching 1 as m and n grow. Moreover, for the indicated values of s and randomly selected matrices A , one has $\beta_s(A) \leq O(1)\sqrt{s}$ with probability approaching one when m, n grow.

- The above results can be considered as a good news. A bad news is, that we do *not* know how to check efficiently, given an s and a sensing matrix A , that the matrix is s -good. Indeed, we know that a necessary and sufficient condition for s -goodness of A is the nullspace property (1.3.5); this, however, does not help, since the quantity $\gamma_s(A)$ is difficult to compute: computing it by definition requires solving $2^s C_s^n$ LP programs (P_v), $v \in V_s$, which is an astronomic number already for moderate n unless s is really small, like 1 or 2. And no alternative efficient way to compute $\gamma_s(A)$ is known.

As a matter of fact, not only we do not know how to check s -goodness efficiently; there still is no efficient recipe allowing to build, given m , an $m \times 2m$ matrix A which is provably s -good for s larger than $O(1)\sqrt{m}$ — a much smaller “level of goodness” then the one ($s = O(1)m$) promised by theory for typical randomly generated matrices.⁸ The “common life” analogy of this pitiful situation would be as follows: you know that with probability at least 0.9, a brick in your wall is made of gold, and at the same time, you do not know how to tell a golden brick from a usual one.⁹

1.3.1.5 Verifiable sufficient conditions for s -goodness

As it was already mentioned, we do not know efficient ways to check s -goodness of a given sensing matrix in the case when s is not really small. The difficulty here is the standard: to certify s -goodness, we should verify (1.3.5), and the most natural way to do it, based on computing $\gamma_s(A)$, is blocked: by definition,

$$\gamma_s(A) = \max_z \{\|z\|_{s,1} : Az = 0, \|z\|_1 \leq 1\} \quad (1.3.9)$$

that is, $\gamma_s(A)$ is the *maximum* of a convex function $\|z\|_{s,1}$ over the convex set $\{z : Az = 0, \|z\|_1 \leq 1\}$. Although both the function and the set are simple, maximizing of convex function over a convex

⁷entries “of order of $1/\sqrt{m}$ ” make the Euclidean norms of columns in $m \times n$ matrix A nearly one, which is the most convenient for Compressed Sensing normalization of A .

⁸Note that the naive algorithm “generate $m \times 2m$ matrices at random until an s -good, with s promised by the theory, matrix is generated” is *not* an efficient recipe, since we do not know how to check s -goodness efficiently.

⁹This phenomenon is met in many other situations. E.g., in 1938 Claude Shannon (1916-2001), “the father of Information Theory,” made (in his M.Sc. Thesis!) a fundamental discovery as follows. Consider a Boolean function of n Boolean variables (i.e., both the function and the variables take values 0 and 1 only); as it is easily seen there are 2^{2^n} function of this type, and every one of them can be computed by a dedicated circuit comprised of “switches” implementing just 3 basic operations AND, OR and NOT (like computing a polynomial can be carried out on a circuit with nodes implementing just two basic operation: addition of reals and their multiplication). The discovery of Shannon was that every Boolean function of n variables can be computed on a circuit with no more than $Cn^{-1}2^n$ switches, where C is an appropriate absolute constant. Moreover, Shannon proved that “nearly all” Boolean functions of n variables require circuits with at least $cn^{-1}2^n$ switches, c being another absolute constant; “nearly all” in this context means that the fraction of “easy to compute” functions (i.e., those computable by circuits with less than $cn^{-1}2^n$ switches) among all Boolean functions of n variables goes to 0 as n goes to ∞ . Now, computing Boolean functions by circuits comprised of switches was an important technical task already in 1938; its role in our today life can hardly be overestimated — the outlined computation is nothing but what is going on in a computer. Given this observation, it is not surprising that the Shannon discovery of 1938 was the subject of countless refinements, extensions, modifications, etc., etc. What is still missing, is a *single individual example* of a “difficult to compute” Boolean function: as a matter of fact, all multivariate Boolean functions $f(x_1, \dots, x_n)$ people managed to describe explicitly are computable by circuits with just *linear* in n number of switches!

set typically is difficult. The only notable exception here is the case of maximizing a convex function f over a convex set X given as the convex hull of a finite set: $X = \text{Conv}\{v^1, \dots, v^N\}$. In this case, a maximizer of f on the finite set $\{v^1, \dots, v^N\}$ (this maximizer can be found by brute force computation of the values of f at v^i) is the maximizer of f over the entire X (check it yourself or see Section C.5).

Given that the nullspace property “as it is” is difficult to check, we can look for “the second best thing” — efficiently computable *upper and lower bounds* on the “goodness” $s_*(A)$ of A (i.e., on the largest s for which A is s -good).

Let us start with efficient lower bounding of $s_*(A)$, that is, with efficiently verifiable *sufficient conditions* for s -goodness. One way to derive such a condition is to specify an efficiently computable *upper bound* $\hat{\gamma}_s(A)$ on $\gamma_s(A)$. With such a bound at our disposal, the efficiently verifiable condition $\hat{\gamma}_s(A) < 1/2$ clearly will be a sufficient condition for the validity of (1.3.5).

The question is, how to find an efficiently computable upper bound on $\gamma_s(A)$, and here is one of the options:

$$\begin{aligned} \gamma_s(A) &= \max_z \left\{ \max_{v \in V_s} v^T z : Az = 0, \|z\|_1 \leq 1 \right\} \\ \Rightarrow \forall H \in \mathbf{R}^{m \times n} : \quad \gamma_s(A) &= \max_z \left\{ \max_{v \in V_s} v^T [1 - H^T A]z : Az = 0, \|z\|_1 \leq 1 \right\} \\ &\leq \max_z \left\{ \max_{v \in V_s} v^T [1 - H^T A]z : \|z\|_1 \leq 1 \right\} \\ &= \max_{z \in Z} \|[I - H^T A]z\|_{s,1}, \quad Z = \{z : \|z\|_1 \leq 1\}. \end{aligned}$$

We see that whatever be “design parameter” $H \in \mathbf{R}^{m \times n}$, the quantity $\gamma_s(A)$ does not exceed the maximum of a convex function $\|[I - H^T A]z\|_{s,1}$ of z over the unit ℓ_1 -ball Z . But the latter set is perfectly well suited for maximizing convex functions: it is the convex hull of a small (just $2n$ points, \pm basic orths) set. We end up with

$$\forall H \in \mathbf{R}^{m \times n} : \gamma_s(A) \leq \max_{z \in Z} \|[I - H^T A]z\|_{s,1} = \max_{1 \leq j \leq n} \|\text{Col}_j[I - H^T A]\|_{s,1},$$

where $\text{Col}_j(B)$ denotes j -th column of a matrix B . We conclude that

$$\gamma_s(A) \leq \hat{\gamma}_s(A) := \min_H \underbrace{\max_j \|\text{Col}_j[I - H^T A]\|_{s,1}}_{\Psi(H)} \tag{1.3.10}$$

The function $\Psi(H)$ is efficiently computable and convex, this is why its minimization can be carried out efficiently. Thus, $\hat{\gamma}_s(A)$ is an *efficiently computable* upper bound on $\gamma_s(A)$.

Some instructive remarks are in order.

1. The trick which led us to $\hat{\gamma}_s(A)$ is applicable to bounding from above the maximum of a convex function f over the set X of the form $\{x \in \text{Conv}\{v^1, \dots, v^N\} : Ax = 0\}$ (i.e., over the intersection of an “easy for convex maximization” domain and a linear subspace. The trick is merely to note that if A is $m \times n$, then for every $H \in \mathbf{R}^{m \times n}$ one has

$$\max_x \left\{ f(x) : x \in \text{Conv}\{v^1, \dots, v^N\}, Ax = 0 \right\} \leq \max_{1 \leq i \leq N} f([I - H^T Ax]v^i) \tag{!}$$

Indeed, a feasible solution x to the left hand side optimization problem can be represented as a convex combination $\sum_i \lambda_i v^i$, and since $Ax = 0$, we have also $x = \sum_i \lambda_i [I - H^T A]v^i$;

since f is convex, we have therefore $f(x) \leq \max_i f([I - H^T A]v^i)$, and (!) follows. Since (!) takes place for every H , we arrive at

$$\max_x \left\{ f(x) : x \in \text{Conv}\{v^1, \dots, v^N\}, Ax = 0 \right\} \leq \hat{\gamma} := \max_{1 \leq i \leq N} f([I - H^T A]v^i),$$

and, same as above, $\hat{\gamma}$ is efficiently computable, provided that f is efficiently computable convex function.

2. The efficiently computable upper bound $\hat{\gamma}_s(A)$ is *polyhedrally representable* — it is the optimal value in an explicit LP program. To derive this problem, we start with important by itself polyhedral representation of the function $\|z\|_{s,1}$:

Lemma 1.3.1 *For every $z \in \mathbf{R}^n$ and integer $s \leq n$, we have*

$$\|z\|_{s,1} = \min_{w,t} \left\{ st + \sum_{i=1}^n w_i : |z_i| \leq t + w_i, 1 \leq i \leq n, w \geq 0 \right\}. \quad (1.3.11)$$

Proof. One way to get (1.3.11) is to note that $\|z\|_{s,1} = \max_{v \in V_s} v^T z = \max_{v \in \text{Conv}(V_s)} v^T z$ and

to verify that the convex hull of the set V_s is exactly the polytope $\mathcal{V}_s = \{v \in \mathbf{R}^n : |v_i| \leq 1 \forall i, \sum_i |v_i| \leq s\}$ (or, which is the same, to verify that the vertices of the latter polytope are exactly the vectors from V_s). With this verification at our disposal, we get

$$\|z\|_{s,1} = \max_v \left\{ v^T z : |v_i| \leq 1 \forall i, \sum_i |v_i| \leq s \right\};$$

applying LP Duality, we get the representation (1.3.11). A shortcoming of the outlined approach is that one indeed should prove that the extreme points of \mathcal{V}_s are exactly the points from V_s ; this is a relatively easy exercise which we strongly recommend to do. We, however, prefer to demonstrate (1.3.11) directly. Indeed, if (w, t) is feasible for (1.3.11), then $|z_i| \leq w_i + t$, whence the sum of the s largest magnitudes of entries in z does not exceed st plus the sum of the corresponding s entries in w , and thus — since w is nonnegative — does not exceed $st + \sum_i w_i$. Thus, the right hand side in (1.3.11) is \geq the left hand side. On the other hand, let $|z_{i_1}| \geq |z_{i_2}| \geq \dots \geq |z_{i_s}|$ are the s largest magnitudes of entries in z (so that i_1, \dots, i_s are distinct from each other), and let $t = |z_{i_s}|$, $w_i = \max[|z_i| - t, 0]$. It is immediately seen that (t, w) is feasible for the right hand side problem in (1.3.11) and that $st + \sum_i w_i = \sum_{j=1}^s |z_{i_j}| = \|z\|_{s,1}$. Thus, the right hand side in (1.3.11) is \leq the left hand side. \square

Lemma 1.3.1 straightforwardly leads to the following polyhedral representation of $\hat{\gamma}_s(A)$:

$$\begin{aligned} \hat{\gamma}_s(A) &:= \min_H \max_j \|\text{Col}_j[I - H^T A]\|_{s,1} \\ &= \min_{H, w^j, t^j, \tau} \left\{ \tau : \begin{array}{l} -w_i^j - t^j \leq [I - H^T A]_{ij} \leq w_i^j + t^j \forall i, j \\ w^j \geq 0 \forall j, st^j + \sum_i w_i^j \leq \tau \forall j \end{array} \right\}. \end{aligned}$$

3. The quantity $\hat{\gamma}_1(A)$ is exactly equal to $\gamma_1(A)$ rather than to be an upper bound on the latter quantity.

Indeed, we have

$$\gamma_1(A) = \max_i \max_z \{ |z_i| : Az = 0, \|z\|_1 \leq 1 \} \underbrace{\max_i \max_z \{ z_i : Az = 0, \|z\|_1 \leq 1 \}}_{\gamma_i}$$

Applying LP Duality, we get

$$\gamma_i = \min_h \|e_i - A^T h\|_\infty, \quad (P_i)$$

where e_i are the standard basic orths in \mathbf{R}^n . Denoting by h^i optimal solutions to the latter problem and setting $H = [H^1, \dots, H^n]$, we get

$$\begin{aligned} \gamma_1(A) &= \max_i \gamma_i = \max_i \|e_i - A^T h^i\|_\infty = \max_{i,j} |[I - A^T h^i]_j| \\ &= \max_{i,j} |[I - A^T H]_{ij}| = \max_{i,j} |[I - H^T A]_{ij}| \\ &= \max_i \|\text{Col}_j[I - H^T A]\|_{1,1} \\ &\geq \hat{\gamma}_1(A); \end{aligned}$$

since the opposite inequality $\gamma_1(A) \leq \hat{\gamma}_1(A)$ definitely holds true, we conclude that

$$\hat{\gamma}_1(A) = \gamma_1(A) = \min_H \max_{i,j} |[I - H^T A]_{ij}|.$$

Observe that an optimal solution H to the latter problem can be found column by column, with j -th column h^j of H being an optimal solution to the LP (P_j) ; this is in a nice contrast with computing $\hat{\gamma}_s(A)$ for $s > 1$, where we should solve a single LP with $O(n^2)$ variables and constraints, which is typically much more time consuming than solving $O(n)$ LP's with $O(n)$ variables and constraints each, as it is the case when computing $\hat{\gamma}_1(A)$.

Observe also that if p, q are positive integers, then for every vector z one has $\|z\|_{pq,1} \leq q\|z\|_{p,1}$, and in particular $\|z\|_{s,1} \leq s\|z\|_{1,1} = s\|z\|_\infty$. It follows that if H is such that $\hat{\gamma}_p(A) = \max_j \|\text{Col}_j[I - H^T A]\|_{p,1}$, then $\hat{\gamma}_{pq}(A) \leq q \max_j \|\text{Col}_j[I - H^T A]\|_{p,1} \leq q\hat{\gamma}_p(A)$. In particular,

$$\hat{\gamma}_s(A) \leq s\hat{\gamma}_1(A),$$

meaning that the easy-to-verify condition

$$\hat{\gamma}_1(A) < \frac{1}{2s}$$

is sufficient for the validity of the condition

$$\hat{\gamma}_s(A) < 1/2$$

and thus is sufficient for s -goodness of A .

4. Assume that A and s are such that s -goodness of A can be certified via our verifiable sufficient condition, that is, we can point out an $m \times n$ matrix H such that

$$\gamma := \max_j \|\text{Col}_j[I - H^T A]\|_{s,1} < 1/2.$$

Now, for every $n \times n$ matrix B , any norm $\|\cdot\|$ on \mathbf{R}^n and every vector $z \in \mathbf{R}^n$ we clearly have

$$\|Bz\| \leq \left[\max_j \|\text{Col}_j[B]\| \right] \|z\|_1$$

(why?) Therefore from the definition of γ , for every vector z we have $\|[I - H^T A]z\|_{s,1} \leq \gamma\|z\|_1$, so that

$$\|z\|_{s,1} \leq \|H^T Az\|_{s,1} + \|[I - H^T A]z\|_{s,1} \leq \left[s \max_j \|\text{Col}_j[H]\|_2 \right] \|Az\|_2 + \gamma\|z\|_1,$$

meaning that H certifies not only the s -goodness of A , but also an inequality of the form (1.3.7) and thus – the associated error bound (1.3.8) for imperfect ℓ_1 recovery.

Operational Exercise 1.3.1 1. Mutual incoherence. Let A be an $m \times n$ matrix with columns A_j normalized to have Euclidean lengths equal to 1. The quantity

$$\mu(A) = \max_{i \neq j} |A_i^T A_j|$$

is called *mutual incoherence* of A , the smaller is this quantity, the closer the columns of A are to mutual orthogonality.

- (a) Prove that $\gamma_1(A) = \hat{\gamma}_1(A) \leq \frac{\mu(A)}{\mu(A)+1}$ and that whenever $s < \frac{\mu(A)+1}{2\mu(A)}$, the relation (1.3.7) is satisfied with

$$\gamma = \frac{s\mu(A)}{\mu(A)+1} < 1/2, \quad \beta = \frac{s\mu(A)}{\mu(A)+1}.$$

Hint: Look what happens with the verifiable sufficient condition for s -goodness when $H = \frac{\mu(A)}{\mu(A)+1} A$.

- (b) Let A be randomly selected $m \times n$ matrix, $1 \ll m \leq n$, with independent entries taking values $\pm 1/\sqrt{m}$ with probabilities 0.5. Verify that for a properly chosen absolute constant C we have $\mu(A) \leq C\sqrt{\ln(n)/m}$. Derive from this observation that the above verifiable sufficient condition for s -goodness for properly selected $m \times n$ matrices can certify their s -goodness with s as large as $O(\sqrt{m}/\ln(n))$.
2. Limits of performance of verifiable sufficient condition for s -goodness. Demonstrate that when $m \leq n/2$, our verifiable sufficient condition for s -goodness does not allow to certify s -goodness of A with $s \geq C\sqrt{m}$, C being an appropriate absolute constant.
3. Upper-bounding the goodness level. In order to demonstrate that A is not s -good for a given s , it suffices to point out a vector $z \in \text{Ker}A \setminus \{0\}$ such that $\|z\|_1 \leq 1$ and $\|z\|_{s,1} \geq \frac{1}{2}$. The simplest way to attempt to achieve this goal is as follows: Start with a whatever $v^0 \in V_s$ and solve the LP $\max_z \{[v^0]^T z : Az = 0, \|z\|_1 \leq 1\}$, thus getting z^1 such that $\|z^1\|_{s,1} \geq [v^0]^T z^1$ and $Az^1 = 0, \|z^1\|_1 \leq 1$. Now choose $v^1 \in V_s$ such that $[v^1]^T z^1 = \|z^1\|_{s,1}$, and solve the LP program $\max_z \{[v^1]^T z : Az = 0, \|z\|_1 \leq 1\}$, thus getting a new $z = z^2$, then define $v^2 \in V_s$ such that $[v^2]^T z^2 = \|z^2\|_{s,1}$, solve the new LP, and so on. In short, the outlined process is an attempt to lower-bound $\gamma_s(A) = \max_{\substack{v \in V_s \\ z: Az=0, \|z\|_1 \leq 1}} v^T z$ by switching

from maximization in z to maximization in v and vice versa. What we can ensure is that $\|z^t\|_{s,1}$ grows with t and $Az^t = 0, \|z^t\|_1 \leq 1$ for all t . With luck, at certain step we get $\|z^t\|_{s,1} \geq 1/2$, meaning that A is not s -good, and z^t can be converted into an s -sparse signal for which the ℓ_1 recovery does not work properly. Alternatively, the routine eventually “gets stuck:” the norms $\|z^t\|_{s,1}$ nearly do not grow with t , which, practically speaking, means that we have nearly reached a local maximum of the function $\|z\|_{s,1}$ on the set $\{Az = 0, \|z\|_1 \leq 1\}$. This local maximum, however, is not necessarily global, so that getting stuck with the value of $\|z\|_{s,1}$ like 0.4 (or even 0.499999) does not allow us to make any conclusion on whether or not $\gamma_s(A) < 1/2$. In this case it makes sense to restart our procedure from a new randomly selected v^0 , and run this process with restarts during the time we are ready to spend on upper-bounding of the goodness level.

4. Running experiment. Use `cvx` to run experiment as follows:

- Build 64×64 Hadamard matrix¹⁰, extract from it at random a 56×64 submatrix and scale the columns of the resulting matrix to have Euclidean norms equal to 1; what you get is your 56×64 sensing matrix A .
- Compute the best lower bounds on the level $s_*(A)$ of goodness of A as given by mutual incoherence, our simplified sufficient condition “ A is s -good for every s for which $s\hat{\gamma}_1(A) < 1/2$,” and our “full strength” verifiable sufficient condition “ A is s -good for every s such that $\hat{\gamma}_s(A) < 1/2$.”
- Compute an upper bound on $s_*(A)$. How far off is this bound from the lower one?
- Finally, take your upper bound on $s_*(A)$, increase it by about 50%, thus getting some S such that A definitely is not S -good, and run 100 experiments where randomly generated signals x with S nonzero entries each are recovered by ℓ_1 minimization from their noiseless observations Ax . How many failures of the ℓ_1 recovery have you observed?

Rerun your experiments with the 64×64 Hadamard matrix replaced with those drawn from 64×64 Rademacher¹¹ and Gaussian¹² matrix ensembles.

1.3.2 Supervised Binary Machine Learning via LP Support Vector Machines

Imagine that we have a source of *feature vectors* — collections x of n measurements representing, e.g., the results of n medical tests taken from patients, and a patient can be affected, or not affected, by a particular illness. “In reality,” these feature vectors x go along with *labels* y taking values ± 1 ; in our example, the label -1 says that the patient whose test results are recorded in the feature vector x does *not* have the illness in question, while the label $+1$ means that the patient is ill.

We assume that there is certain dependence between the feature vectors and the labels, and our goal is to predict, given a feature vector alone, the value of the label. What we have at our disposal is a *training sample* (x^i, y^i) , $1 \leq i \leq N$ of examples (x^i, y^i) where we know both the

¹⁰Hadamard matrices are square $2^k \times 2^k$ matrices H_k , $k = 0, 1, 2, \dots$ given by the recurrence $H_0 = 1$, $H_{k+1} = \begin{bmatrix} H_k & H_k \\ H_k & -H_k \end{bmatrix}$. These are symmetric matrices with ± 1 with columns orthogonal to each other.

¹¹random matrix with independent entries taking values ± 1 with probabilities 1/2.

¹²random matrix with independent standard (zero mean, unit variance) Gaussian entries.

feature vector and the label; given this sample, we want to build a *classifier* – a function $f(x)$ on the space of feature vectors x taking values ± 1 – which we intend to use to predict, given the value of a new feature vector, the value of the corresponding label. In our example this setup reads: we are given medical records containing both the results of medical tests and the diagnoses of N patients; given this data, we want to learn how to predict the diagnosis given the results of the tests taken from a new patient.

The simplest predictors we can think about are just the “linear” ones looking as follows. We fix an affine form $z^T x + b$ of a feature vector, choose a positive threshold γ and say that if the value of the form at a feature vector x is “well positive” – is $\geq \gamma$ – then the proposed label for x is $+1$; similarly, if the value of the form at x is “well negative” – is $\leq -\gamma$, then the proposed label will be -1 . In the “gray area” $-\gamma < z^T x + b < \gamma$ we decline to classify. Noting that the actual value of the threshold is of no importance (to compensate a change in the threshold by certain factor, it suffices to multiply by this factor both z and b , without affecting the resulting classification), we from now on normalize the situation by setting the threshold to the value 1.

Now, we have explained how a linear classifier works, but where from to take it? An intuitively appealing idea is to use the training sample in order to “train” our potential classifier – to choose z and b in a way which ensures correct classification of the examples in the sample. This amounts to solving the system of linear inequalities

$$z^T x^i + b \geq 1 \quad \forall (i \leq N : y^i = +1) \quad \& \quad z^T x^i + b \leq -1 \quad \forall (i : y^i = -1),$$

which can be written equivalently as

$$y^i(z_i^T x^i + b) \geq 1 \quad \forall i = 1, \dots, N.$$

Geometrically speaking, we want to find a “stripe”

$$-1 < z^T x + b < 1 \tag{*}$$

between two parallel hyperplanes $\{x : z^T x + b = -1\}$ and $\{x : z^T x + b = 1\}$ such that all “positive examples” (those with the label $+1$) from the training sample are on one side of this stripe, while all negative (the label -1) examples from the sample are on the other side of the stripe. With this approach, it is natural to look for the “thickest” stripe separating the positive and the negative examples. Since the geometric width of the stripe is $\frac{2}{\sqrt{z^T z}}$ (why?), this amounts to solving the optimization program

$$\min_{z,b} \left\{ \|z\|_2 := \sqrt{z^T z} : y^i(z^T x^i + b) \geq 1, 1 \leq i \leq N \right\}; \tag{1.3.12}$$

The latter problem, of course, not necessarily is feasible: it well can happen that it is impossible to separate the positive and the negative examples in the training sample by a stripe between two parallel hyperplanes. To handle this possibility, we allow for classification errors and minimize a weighted sum of $\|w\|_2$ and total penalty for these errors. Since the absence of classification penalty at an example (x^i, y^i) in outer context is equivalent to the validity of the inequality $y^i(w^T x^i + b) \geq 1$, the most natural penalty for misclassification of the example is $\max[1 - y^i(z^T x^i + b), 0]$. With this in mind, the problem of building “the best on the training sample” classifier becomes the optimization problem

$$\min_{z,b} \left\{ \|z\|_2 + \lambda \sum_{i=1}^N \max[1 - y^i(z^T x^i + b), 0] \right\}, \tag{1.3.13}$$

where $\lambda > 0$ is responsible for the “compromise” between the width of the stripe (*) and the “separation quality” of this stripe; how to choose the value of this parameter, this is an additional story we do not touch here. Note that the outlined approach to building classifiers is the most basic and the most simplistic version of what in Machine Learning is called “Support Vector Machines.”

Now, (1.3.13) is *not* an LO program: we know how to get rid of nonlinearities $\max[1 - y^i(w^T x^i + b), 0]$ by adding slack variables and linear constraints, but we cannot get rid of the nonlinearity brought by the term $\|z\|_2$. Well, there are situations in Machine Learning where it makes sense to get rid of this term by “brute force,” specifically, by replacing the $\|\cdot\|_2$ with $\|\cdot\|_1$. The rationale behind this “brute force” action is as follows. The dimension n of the feature vectors can be large, In our medical example, it could be in the range of tens, which perhaps is “not large;” but think about digitalized images of handwritten letters, where we want to distinguish between handwritten letters “A” and “B;” here the dimension of x can well be in the range of thousands, if not millions. Now, it would be highly desirable to design a good classifier with *sparse* vector of weights z , and there are several reasons for this desire. First, intuition says that a good on the training sample classifier which takes into account just 3 of the features should be more “robust” than a classifier which ensures equally good classification of the training examples, but uses for this purpose 10,000 features; we have all reasons to believe that the first classifier indeed “goes to the point,” while the second one adjusts itself to random, irrelevant for the “true classification,” properties of the training sample. Second, to have a good classifier which uses small number of features is definitely better than to have an equally good classifier which uses a large number of them (in our medical example: the “predictive power” being equal, we definitely would prefer predicting diagnosis via the results of 3 tests to predicting via the results of 20 tests). Finally, if it is possible to classify well via a small number of features, we hopefully have good chances to *understand the mechanism of the dependencies between these measured features and the feature which presence/absence we intend to predict* — it usually is much easier to understand interaction between 2-3 features than between 2,000-3,000 of them. Now, the SVMs (1.3.12), (1.3.13) are not well suited for carrying out the outlined *feature selection* task, since minimizing $\|z\|_2$ norm under constraints on z (this is what explicitly goes on in (1.3.12) and implicitly goes on in (1.3.13)¹³) typically results in “spread” optimal solution, with many small nonzero components. In view of our “Compressed Sensing” discussion, we could expect that minimizing the ℓ_1 -norm of z will result in “better concentrated” optimal solution, which leads us to what is called “LO Support Vector Machine.” Here the classifier is given by the solution of the $\|\cdot\|_1$ -analogy of (1.3.13), specifically, the optimization problem

$$\min_{z,b} \left\{ \|z\|_1 + \lambda \sum_{i=1}^N \max[1 - y^i(z^T x^i + b), 0] \right\}. \quad (1.3.14)$$

¹³To understand the latter claim, take an optimal solution (z_*, b_*) to (1.3.13), set $\Lambda = \sum_{i=1}^N \max[1 - y^i(z_*^T x^i + b_*), 0]$ and note that (z_*, b_*) solves the optimization problem

$$\min_{z,b} \left\{ \|z\|_2 : \sum_{i=1}^N \max[1 - y^i(z^T x^i + b), 0] \leq \Lambda \right\}$$

(why?).

This problem clearly reduces to the LO program

$$\min_{z,b,w,\xi} \left\{ \sum_{j=1}^n w_j + \lambda \sum_{i=1}^N \xi_i : -w_j \leq z_j \leq w_j, 1 \leq j \leq n, \xi_i \geq 0, \xi_i \geq 1 - y^i(z^T x^i + b), 1 \leq i \leq N \right\}. \quad (1.3.15)$$

Concluding remarks. A reader could ask, what is the purpose of training the classifier on the training set of examples, where we from the very beginning know the labels of all the examples? why a classifier which classifies well on the training set should be good at new examples? Well, intuition says that if a simple rule with a relatively small number of “tuning parameters” (as it is the case with a sparse linear classifier) recovers well the labels in examples from a large enough sample, this classifier should have learned something essential about the dependency between feature vectors and labels, and thus should be able to classify well new examples. Machine Learning theory offers a solid probabilistic framework in which “our intuition is right”, so that under assumptions (not too restrictive) imposed by this framework it is possible to establish quantitative links between the size of the training sample, the behavior of the classifier on this sample (quantified by the $\|\cdot\|_2$ or $\|\cdot\|_1$ norm of the resulting z and the value of the penalty for misclassification), and the *predictive power* of the classifier, quantified by the probability of misclassification of a new example; roughly speaking, good behavior of a linear classifier achieved at a large training sample ensures low probability of misclassifying a new example.

1.3.3 Synthesis of linear controllers

1.3.3.1 Discrete time linear dynamical systems

The most basic and well studied entity in control is a *linear dynamical system* (LDS). In the sequel, we focus on *discrete time LDS* modeled as

$$\begin{aligned} x_0 &= z && [\text{initial condition}] \\ x_{t+1} &= A_t x_t + B_t u_t + R_t d_t, && [\text{state equations}] \\ y_t &= C_t x_t + D_t d_t && [\text{outputs}] \end{aligned} \quad (1.3.16)$$

In this description, $t = 0, 1, 2, \dots$ are time instants,

- $x_t \in \mathbf{R}^{n_x}$ is *state* of the system at instant t ,
- $u_t \in \mathbf{R}^{n_u}$ is *control* generated by system’s controller at instant t ,
- $d_t \in \mathbf{R}^{n_d}$ is *external disturbance* coming from system’s environment at instant t ,
- $y_t \in \mathbf{R}^{n_y}$ is *observed output* at instant t ,
- A_t, B_t, \dots, D_t are (perhaps depending on t) matrices of appropriate sizes specifying system’s dynamics and relations between states, controls, external disturbances and outputs.

What we have described so far is called an *open loop* system (or open loop plant). This plant should be augmented by a *controller* which generates subsequent controls. The standard assumption is that the control u_t is generated in a deterministic fashion and depends on the outputs y_1, \dots, y_t observed prior to instant t and at this very instant (*non-anticipative*, or *causal* control):

$$u_t = U_t(y_0, \dots, y_t); \quad (1.3.17)$$

here U_t are arbitrary everywhere defined functions of their arguments taking values in \mathbf{R}^{n_u} . Plant augmented by controller is called a *closed loop* system; its behavior clearly depends on the initial state and external disturbances only.

Affine control. The simplest (and extremely widely used) form of control law is *affine* control, where u_t are affine functions of the outputs:

$$u_t = \xi_t + \Xi_0^t y_0 + \Xi_1^t y_1 + \dots + \Xi_t^t y_t, \quad (1.3.18)$$

where ξ_t are vectors, and Ξ_τ^t , $0 \leq \tau \leq t$, are matrices of appropriate sizes.

Augmenting linear open loop system (1.3.16) with affine controller (1.3.18), we get a well defined closed loop system in which states, controls and outputs are *affine* functions of the initial state and external disturbances; moreover, x_t depends solely on the initial state z and the collection $d^{t-1} = [d_0; \dots; d_{t-1}]$ of disturbances prior to instant t , while u_t and y_t may depend on z and the collection $d^t = [d_0; \dots; d_t]$ of disturbances including the one at instant t .

Design specifications and the Analysis problem. The entities of primary interest in control are states and controls; we can arrange states and controls into a long vector — the *state-control trajectory*

$$w^N = [x_N; x_{N-1}; \dots; x_1; u_{N-1}; \dots; u_0];$$

here N is the *time horizon* on which we are interested in system's behavior. with affine control law (1.3.18), this trajectory is an *affine* function of z and d^N :

$$w^N = w^N(z, d^N) = \omega_N + \Omega_N[z; d^N]$$

with vector ω_N and matrix Ω_N are readily given by the matrices A_t, \dots, D_t , $0 \leq t < N$, from the description of the open loop system and by the collection $\tilde{\xi}^N = \{\xi_t, \Xi_\tau^t : 0 \leq \tau \leq t < N\}$ of the parameters of the affine control law (1.3.18).

Imagine that the “desired behaviour” of the closed loop system on the time horizon in question is given by a system of *linear* inequalities

$$\mathcal{B}w^N \leq b \quad (1.3.19)$$

which should be satisfied by the state-control trajectory provided that the initial state z and the disturbances d^N vary in their “normal ranges” \mathcal{Z} and \mathcal{D}^N , respectively; this is a pretty general form of design specifications. The fact that w^N depends *affinely* on $[z; d^N]$ makes it easy to solve the *Analysis problem*: to check whether a given control law (1.3.18) ensures the validity of design specifications (1.3.19). Indeed, to this end we should check whether the functions $[\mathcal{B}w^N - b]_i$, $1 \leq i \leq I$ (I is the number of linear inequalities in (1.3.19)) remain nonpositive whenever $d^N \in \mathcal{D}^N$, $z \in \mathcal{Z}$; for a given control law, the functions in question are explicitly given *affine* function $\phi_i(z, d^N)$ of $[z; d^N]$, so that what we need to verify is that

$$\max_{[z; d^N]} \left\{ \phi_i(z, d^N) : [z; d^N] \in \mathcal{Z}\mathcal{D}^N := \mathcal{Z} \times \mathcal{D}^N \right\} \leq 0, \quad i = 1, \dots, I.$$

Whenever \mathcal{D}^N and \mathcal{Z} are explicitly given convex sets, the latter problems are convex and thus easy to solve. Moreover, if $\mathcal{Z}\mathcal{D}^N$ is given by *polyhedral representation*

$$\mathcal{Z}\mathcal{D}^N := \mathcal{D}^N \times \mathcal{Z} = \left\{ [z; d^N] : \exists v : P[z; d^N] + Qv \leq r \right\} \quad (1.3.20)$$

(this is a pretty flexible and enough general way to describe typical ranges of disturbances and initial states), the analysis problem reduces to a bunch of explicit LPs

$$\max_{z, d^N, v} \left\{ \phi_i([z; d^N]) : P[z; d^N] + Qv \leq r \right\}, \quad 1 \leq i \leq I;$$

the answer in the analysis problem is positive if and only if the optimal values in all these LPs are nonpositive.

Synthesis problem. As we have seen, with affine control and affine design specifications, it is easy to check whether a given control law meets the specifications. The basic problem in linear control is, however, somehow different: usually we need to *build* an affine control which meets the design specifications (or to detect that no such control exists). And here we run into difficult problem: while state-control trajectory for a given affine control is an easy-to-describe *affine* function of $[z; d^N]$, its dependence on the collection ξ^N of the parameters of the control law is highly nonlinear, even for a time-invariant system (the matrices A_t, \dots, D_t are independent of t) and a control law as simple as time-invariant linear feedback: $u_t = Ky_t$. Indeed, due to the dynamic nature of the system, in the expressions for states and controls powers of the matrix K will be present. Highly nonlinear dependence of states and controls on S makes it impossible to optimize efficiently w.r.t. the parameters of the control law and thus makes the synthesis problem extremely difficult.

The situation, however, is far from being hopeless. We are about to demonstrate that *one can re-parameterize affine control law in such a way that with the new parameterization both Analysis and Synthesis problems become tractable*.

1.3.3.2 Purified outputs and purified-output-based control laws

Imagine that we “close” the open loop system with a whatever (affine or non-affine) control law (1.3.17) and in parallel with running the closed loop system run its *model*:

	System:	
	$x_0 = z$	
	$x_{t+1} = A_t x_t + B_t u_t + R_t d_t,$	
	$y_t = C_t x_t + D_t d_t$	
<hr/>		
	Model:	
	$\hat{x}_0 = 0$	(1.3.21)
	$\hat{x}_{t+1} = A_t \hat{x}_t + B_t u_t,$	
	$\hat{y}_t = C_t \hat{x}_t$	
<hr/>		
	Controller:	
	$u_t = U_t(y_0, \dots, y_t)$	(!)

Assuming that we know the matrices A_t, \dots, D_t , we can run the model in an on-line fashion, so that at instant t , when the control u_t should be specified, we have at our disposal both the actual outputs y_0, \dots, y_t and the model outputs $\hat{y}_0, \dots, \hat{y}_t$, and thus have at our disposal the *purified outputs*

$$v_\tau = y_\tau - \hat{y}_\tau, \quad 0 \leq \tau \leq t.$$

Now let us ask ourselves what will happen if we, instead of building the controls u_t on the basis of actual outputs y_τ , $0 \leq \tau \leq t$, pass to controls u_t built on the basis of purified outputs v_τ ,

$0 \leq \tau \leq t$, i.e., replace the control law (!) with control law of the form

$$u_t = V_t(v_0, \dots, v_t) \quad (\text{!!})$$

It easily seen that nothing will happen:

For every control law $\{U_t(y_0, \dots, y_t)\}_{t=0}^{\infty}$ of the form (!) there exists a control law $\{V_t(v_0, \dots, v_t)\}_{t=0}^{\infty}$ of the form (!!) (and vice versa, for every control law of the form (!!)) *there exists a control law of the form (!) such that the dependencies of actual states, outputs and controls on the disturbances and the initial state for both control laws in question are exactly the same.* Moreover, the above “equivalence claim” remains valid when we restrict controls (!), (!!)) to be affine in their arguments.

The bottom line is that every behavior of the close loop system which can be obtained with affine non-anticipative control (1.3.18) based on actual outputs, can be also obtained with affine non-anticipative control law

$$u_t = \eta_t + H_0^t v_0 + H_1^t v_1 + \dots + H_t^t v_t, \quad (1.3.22)$$

based on purified outputs (and vice versa).

Exercise 1.1 Justify the latter conclusion.

We have said that as far as achievable behaviors of the closed loop system are concerned, we loose (and gain) nothing when passing from affine output-based control laws (1.3.18) to affine purified output-based control laws (1.3.22). At the same time, when passing to purified outputs, we get a huge bonus:

(#) With control (1.3.22), the trajectory w^N of the closed loop system turns out to be bi-affine: it is affine in $[z; d^N]$, the parameters $\vec{\eta}^N = \{\eta_t, H_\tau^t, 0 \leq \tau \leq t \leq N\}$ of the control law being fixed, and is affine in the parameters of the control law $\vec{\eta}^N$, $[z; d^N]$ being fixed, and this bi-affinity, as we shall see in a while, is the key to efficient solvability of the synthesis problem.

The reason for bi-affinity is as follows (after this reason is explained, verification of bi-affinity itself becomes immediate): purified output v_t is completely independent on the controls and is a known in advance affine function of d^t, z . Indeed, from (1.3.21) it follows that

$$v_t = C_t \underbrace{(x_t - \hat{x}_t)}_{\delta_t} + D_t d_t$$

and that the evolution of δ_t is given by

$$\delta_0 = z, \delta_{t+1} = A_t \delta_t + R_t d_t$$

and thus is completely independent of the controls, meaning that δ_t and v_t indeed are known in advance (provided the matrices A_t, \dots, D_t are known in advance) affine functions of d^t, z .

Exercise 1.2 Derive (#) from the observation we just have made.

Note that in contrast to the just outlined “control-independent” nature of purified outputs, the actual outputs are heavily control-dependent (indeed, y_t is affected by x_t , and this state, in turn, is affected by past controls u_0, \dots, u_{t-1}). This is why with the usual output-based affine control, the states and the controls are highly nonlinear in the parameters of the control law – to build u_i , we multiply matrices Ξ_τ^t (which are parameters of the control law) by outputs y_τ which by themselves already depend on the “past” parameters of the control law.

Tractability of the Synthesis problem. Assume that the normal range \mathcal{ZD}^N of $[z; d^N]$ is a nonempty and bounded set given by polyhedral representation (1.3.20), and let us prove that in this case the design specifications (1.3.19) reduce to a system of explicit linear inequalities in variables $\vec{\eta}^N$ – the parameters of the purified-output-based affine control law used to close the open-loop system – and appropriate slack variables. Thus, (parameters of) purified-output-based affine control laws meeting the design specifications (1.3.19) form a polyhedrally representable, and thus easy to work with, set.

The reasoning goes as follows. As stated by (#), the state-control trajectory w^N associated with affine purified-output-based control with parameters $\vec{\eta}^N$ is bi-affine in d^N, z and in $\vec{\eta}^N$, so that it can be represented in the form

$$w^N = w[\vec{\eta}^N] + W[\vec{\eta}^N] \cdot [z; d^N],$$

where the vector-valued and the matrix-valued functions $w[\cdot], W[\cdot]$ are affine and are readily given by the matrices $A_t, \dots, D_t, 0 \leq t \leq N - 1$. Plugging in the representation of w^N into the design specifications (1.3.19), we get a system of scalar constraints of the form

$$\alpha_i^T(\vec{\eta}^N)[z; d^N] \leq \beta_i(\vec{\eta}^N), \quad 1 \leq i \leq I, \quad (*)$$

where the vector-valued functions $\alpha_i(\cdot)$ and the scalar functions $\beta_i(\cdot)$ are affine and readily given by the description of the open-loop system and by the data \mathcal{B}, b in the design specifications. What we want from $\vec{\eta}^N$ is to ensure the validity of every one of the constraints (*) for all $[z; d^N]$ from \mathcal{ZD}^n , or, which is the same in view of (1.3.20), we want the optimal values in the LPs

$$\max_{[z; d^N], v} \left\{ \alpha_i^T(\vec{\eta}^N)[z; d^N] : P[z; d^N] + Qv \leq r \right\}$$

to be $\leq \beta_i(\vec{\eta}^N)$ for $1 \leq i \leq I$. Now, the LPs in question are feasible; passing to their duals, what we want become exactly the relations

$$\min_{s_i} \left\{ r^T s_i : P^T s_i = \beta_i(\vec{\eta}^N), Q^T s_i = 0, s_i \geq 0 \right\} \leq \beta_i(\vec{\eta}^N), \quad 1 \leq i \leq I.$$

The bottom line is that

A purified-output-based control law meets the design specifications (1.3.19) if and only if the corresponding collection $\vec{\eta}^N$ of parameters can be augmented by properly chosen slack vector variables s_i to give a solution to the system of linear inequalities in variables $s_1, \dots, s_I, \vec{\kappa}^N$, specifically, the system

$$P^T s_i = \beta_i(\vec{\eta}^N), \quad Q^T s_i = 0, \quad r^T s_i \leq \beta_i(\vec{\eta}^N), \quad s_i \geq 0, \quad 1 \leq i \leq I. \quad (1.3.23)$$

Remark 1.3.1 We can say that when passing from an affine output-based control laws to affine purifies output based ones we are all the time dealing with the same entities (affine non-anticipative control laws), but switch from one parameterization of these laws (the one by $\vec{\xi}^N$ -parameters) to another parameterization (the one by $\vec{\eta}^N$ -parameters). This re-parameterization is nonlinear, so that in principle there is no surprise that what is difficult in one of them (the Synthesis problem) is easy in another one. This being said, note that with our re-parameterization we neither lose nor gain only as far as the entire family of linear controllers is concerned. Specific sub-families of controllers can be “simply-looking” in one of the parameterizations and be extremely difficult to describe in the other one. For example, time-invariant linear feedback

$u_t = Ky_t$ looks pretty simple (just a linear subspace) in the ξ^N -parameterization and form a highly nonlinear manifold in the $\bar{\eta}^N$ -one. Similarly, the a.p.o.b. control $u_t = Kv_t$ looks simple in the $\bar{\eta}^N$ -parameterization and is difficult to describe in the ξ^N -one. We can say that there is no such thing as “the best” parameterization of affine control laws — everything depends on what are our goals. For example, the $\bar{\eta}^N$ parameterization is well suited for synthesis of general-type affine controllers and becomes nearly useless when a linear feedback is sought.

Modifications. We have seen that if the “normal range” \mathcal{ZD}^N of $[z; d^N]$ is given by polyhedral representation (which we assume from now on), the parameters of “good” (i.e., meeting the design specifications (1.3.19) for all $[z; d^N] \in \mathcal{ZD}^N$) affine purified-output-based (a.p.o.b.) control laws admit an explicit polyhedral representation — these are exactly the collections $\bar{\eta}^N$ which can be extended by properly selected slacks s_1, \dots, s_N to a solution of the explicit system of linear inequalities (1.3.23). As a result, various synthesis problems for a.p.o.b. controllers become just explicit LPs. This is the case, e.g., for the problem of checking whether the design specifications can be met (this is just a LP feasibility problem of checking solvability of (1.3.23)). If this is the case, we can further minimize efficiently a “good” (i.e., convex and efficiently computable) objective over the a.p.o.b. controllers meeting the design specifications; when the objective is just linear, this will be just an LP program.

In fact, our approach allows for solving efficiently other meaningful problems of a.p.o.b. controller design, e.g., as follows. Above, we have assumed that the range of disturbances and initial states is known, and we want to achieve “good behavior” of the closed loop system in this range; what happens when the disturbances and the initial states go beyond their normal range, is not our responsibility. In some cases, it makes sense to take care of the disturbances/initial states beyond their normal range, specifically, to require the validity of design specifications (1.3.19) in the normal range of disturbances/initial states, and to allow for controlled deterioration of these specifications when the disturbances/initial states run beyond their normal range. This can be modeled as follows: Let us fix a norm $\|\cdot\|$ on the linear space $\mathcal{L}^N = \{\zeta = [z; d^N] \in \mathbf{R}^{n_d} \times \dots \times \mathbf{R}^{n_d} \times \mathbf{R}^{n_x}\}$ where the disturbances and the initial states live, and let

$$\text{dist}([z; d^N], \mathcal{ZD}^N) = \min_{\zeta \in \mathcal{ZD}^N} \| [z; d^N] - \zeta \|$$

be the deviation of $[z; d^N]$ from the normal range of this vector. A natural extension of our previous design specifications is to require the validity of the relations

$$\mathcal{B}w^N(d^N, z) \leq b + \text{dist}([z; d^N], \mathcal{ZD}^N)\alpha, \quad (1.3.24)$$

for all $[z; d^N] \in \mathcal{L}^N$; here, as above, $w^N(d^N, z)$ is the state-control trajectory of the closed loop system on time horizon N treated as a function of $[z; d^N]$, and $\alpha = [\alpha_1; \dots; \alpha_I]$ is a nonnegative vector of sensitivities, which for the time being we consider as given to us in advance. In other words, we still want the validity of (1.3.19) in the normal range of disturbances and initial states, and on the top of it, impose additional restrictions on the global behaviour of the closed loop system, specifically, want the violations of the design specifications for $[z; d^N]$ outside of its normal range to be bounded by given multiples of the distance from $[z; d^N]$ to this range.

As before, we assume that the normal range \mathcal{ZD}^N of $[z; d^N]$ is a nonempty and *bounded* set given by polyhedral representation (1.3.20). We are about to demonstrate that in this case the specifications (1.3.24) still are tractable (and even reduce to LP when $\|\cdot\|$ is a simple enough norm). Indeed, let us look at i -th constraint in (1.3.24). Due to bi-affinity of w^N in $[z; d^N]$ and

in the parameters $\vec{\eta}^N$ of the a.p.o.b. control law in question, this constraint can be written as

$$\alpha_i^T(\vec{\eta}^N)[z; d^N] \leq \beta_i(\vec{\eta}^N) + \alpha_i \text{dist}([z; d^N], \mathcal{ZD}^N) \quad \forall [z; d^N] \in \mathcal{L}^N. \quad (1.3.25)$$

where $\alpha_i(\cdot)$, $\beta_i(\cdot)$ are affine functions. We claim that

(&) Relation (1.3.25) is equivalent to the following two relations:

$$(a) \quad \alpha_i^T(\vec{\eta}^N)[z; d^N] \leq \beta_i(\vec{\eta}^N) \quad \forall [z; d^N]$$

$$(b) \quad f_i(\vec{\eta}^N) := \max_{[z; d^N]} \left\{ \alpha_i^T(\vec{\eta}^N)[z; d^N] : \| [z; d^N] \| \leq 1 \right\} \leq \alpha_i.$$

Postponing for the moment the verification of this claim, let us look at the consequences. We already know that (a) “admits polyhedral representation:” we can point out a system \mathcal{S}_i of linear inequalities on $\vec{\eta}^N$ and slack vector s_i such that (a) takes place if and only if $\vec{\kappa}^N$ can be extended, by a properly chosen s_i , to a solution of \mathcal{S}_i . Further, the function $f_i(\cdot)$ clearly is convex (since $\alpha_i(\cdot)$ is affine) and efficiently computable, provided that the norm $\|\cdot\|$ is efficiently computable (indeed, in this case computing f_i at a point reduces to maximizing a linear form over a convex set). Thus, not only (a), but (b) as well is a tractable constraint on $\vec{\kappa}^N$.

In particular, when $\|\cdot\|$ is a polyhedral norm, meaning that the set $\{[z; d^N] : \| [z; d^N] \| \leq 1\}$ admits polyhedral representation:

$$\{[z; d^N] : \| [z; d^N] \| \leq 1\} = \{[z; d^N] : \exists y : U \cdot [z; d^N] + Yy \leq q\},$$

we have

$$\begin{aligned} f_i(\vec{\eta}^N) &= \max_{[z; d^N], y} \left\{ \alpha_i^T(\vec{\eta}^N)[z; d^N] : U \cdot [z; d^N] + Yy \leq q \right\} \\ &= \min_{t_i} \left\{ q^T t_i : U^T t_i = \alpha_i^T(\vec{\eta}^N), Y^T t_i = 0, t_i \geq 0 \right\} \quad [\text{by LP Duality Theorem}], \end{aligned}$$

We arrive at an explicit “polyhedral reformulation” of (b): $\vec{\eta}^N$ satisfies (b) if and only if $\vec{\eta}^N$ can be extended, by a properly chosen slack vector t_i , to a solution of an explicit system of linear inequalities in variables $\vec{\eta}^N, t_i$, specifically, the system

$$q^T t_i \leq \alpha_i \quad U^T t_i = \alpha_i^T(\vec{\eta}^N), \quad Y^T t_i = 0, \quad t_i \geq 0 \quad (\mathcal{T}_i)$$

Thus, for a polyhedral norm $\|\cdot\|$, (a), (b) can be expressed by a system of linear inequalities in the parameters $\vec{\eta}^N$ of the a.p.o.b. control law we are interested in and in slack variables s_i, t_i .

Remark 1.3.2 For the time being, we treated sensitivities α_i as given parameters. Our analysis, however, shows that in “tractable reformulation” of (1.3.25) α_i are just the right hand sides of some inequality constraints, meaning that the nice structure of these constraints (convexity/linearity) is preserved when treating α_i as variables (restricted to be ≥ 0) rather than given parameters. As a result, we can solve efficiently the problem of synthesis an a.p.o.b. controller which ensures (1.3.25) and minimizes one of α_i ’s (or some weighted sum of α_i ’s) under additional, say, linear, constraints on α_i ’s, and there are situations when such a goal makes perfect sense.

Clearing debts. It remains to prove (b), which is easy. Leaving the rest of the justification of (b) to the reader, let us focus on the only not completely evident part of the claim, namely, that (1.3.25) implies (b). The reasoning is as follows: take a vector $[\bar{z}; \bar{d}^N] \in \mathcal{L}^N$ with $\|[\bar{z}; \bar{d}^N]\| \leq 1$ and a vector $[\hat{z}; \hat{d}^N] \in \mathcal{ZD}^N$, and let $t > 0$. (1.3.25) should hold for $[z; d^N = [\hat{z}; \hat{d}^N] + t[\bar{z}; \bar{d}^N]]$, meaning that

$$\alpha_i^T(\vec{\eta}^N) ([\hat{z}; \hat{d}^N] + t[\bar{z}; \bar{d}^N]) \leq \beta_i(\vec{\eta}^N) + \alpha_i \underbrace{\text{dist}([z; d^N], \mathcal{ZD}^N)}_{\leq t \| [z; d^N] - [\hat{z}; \hat{d}^N] \| = t \| [\bar{z}; \bar{d}^N] \| \leq t}$$

Dividing both sides by t and passing to limit as $t \rightarrow \infty$, we get

$$\alpha_i^T(\vec{\eta}^N) [\bar{z}; \bar{d}^N] \leq \alpha_i,$$

and this is so for every $[\bar{z}; \bar{d}^N]$ of $\|\cdot\|$ -norm not exceeding 1; this is exactly (b).

Operational Exercise 1.3.2 The goal of this exercise is to play with the outlined synthesis of linear controllers via LP.

1. *Situation and notation.* From now on, we consider discrete time linear dynamic system (1.3.16) which is *time-invariant*, meaning that the matrices A_t, \dots, D_t are independent of t (so that we write A, \dots, D instead of A_t, \dots, D_t). As always, N is the time horizon.

Let us start with convenient notation. Affine p.o.b. control law $\vec{\eta}^N$ on time horizon N can be represented by a pair comprised of the block-vector $\vec{h} = [h_0; h_1; \dots, h_{N-1}]$ and the block-lower-triangular matrix

$$\mathbf{H} = \begin{bmatrix} H_0^0 & & & & \\ H_0^1 & H_1^1 & & & \\ H_0^2 & H_1^2 & H_2^2 & & \\ \vdots & \vdots & \vdots & \ddots & \\ H_0^{N-1} & \dots & \dots & \dots & H_{N-1}^{N-1} \end{bmatrix}$$

with $n_u \times n_y$ blocks H_τ^t (from now on, when drawing a matrix, blanks represent zero entries/blocks). Further, we can stack in long vectors other entities of interest, specifically,

$$w^N := \begin{bmatrix} \vec{x} \\ \vec{u} \end{bmatrix} = \begin{bmatrix} x_1 \\ \vdots \\ x_N \\ u_0 \\ \vdots \\ u_{N-1} \end{bmatrix} \quad \text{[state-control trajectory]}$$

$$\vec{v} = \begin{bmatrix} v_0 \\ \vdots \\ v_{N-1} \end{bmatrix} \quad \text{[purified outputs]}$$

and introduce block matrices

$$\begin{aligned}\zeta 2v &= \begin{bmatrix} C & D \\ CA & CR \\ CA^2 & CAR \\ \vdots & \vdots \\ CA^{N-1} & CA^{N-2}R \\ A & R \\ A^2 & AR \\ A^3 & A^2R \\ \vdots & \vdots \\ A^N & A^{N-1}R \end{bmatrix} \quad \left[\text{Note that } \vec{v} = \zeta 2v \cdot [z; d^N] \right] \\ \zeta 2x &= \begin{bmatrix} D \\ CR \\ CAR \\ \vdots \\ CA^{N-3}R \\ R \\ A^2R \\ AR \\ \vdots \\ A^{N-2}R \\ R \end{bmatrix} \quad \left[\text{Note that the contribution of } [z; d^N] \text{ to } \vec{x} \text{ for a given } \vec{u} \text{ is } \zeta 2x \cdot [z; d^N] \right] \\ u2x &= \begin{bmatrix} B \\ AB \\ A^2B \\ \vdots \\ A^{N-1}B \end{bmatrix} \quad \left[\text{Note that the contribution of } \vec{u} \text{ to } \vec{x} \text{ is } u2x \cdot \vec{u} \right]\end{aligned}$$

In this notation, the dependencies between $[z; d^N]$, $\vec{\eta}^N = (h, H)$, w^N and \vec{v} are given by

$$\begin{aligned}\vec{v} &= \zeta 2v \cdot [z; d^N], \\ w^N := \begin{bmatrix} \vec{x} \\ \vec{u} \end{bmatrix} &= [u2x \cdot \mathbf{H} \cdot \zeta 2v + \zeta 2x] \cdot [z; d^N] + u2x \cdot h.\end{aligned}\quad (1.3.26)$$

2. Task 1:

- (a) Verify (1.3.26).
- (b) Verify that the system closed by an a.p.o.b. control law $\vec{\eta}^N = (\vec{h}, \mathbf{H})$ satisfies the specifications

$$\mathcal{B}w^N \leq b$$

whenever $[z; d^N]$ belongs to a nonempty set given by polyhedral representation

$$\{[z; d^N] : \exists u : P[z; d^N] + Qu \leq r\}$$

if and only if there exists entrywise nonnegative matrix Λ such that

$$\begin{aligned}\Lambda P &= \mathcal{B} \cdot \begin{bmatrix} u2x \cdot \mathbf{H} \cdot \zeta 2v + \zeta 2x \\ \mathbf{H} \cdot \zeta 2v \end{bmatrix} \\ \Lambda Q &= 0 \\ \Lambda r + \mathcal{B} \cdot \begin{bmatrix} u2x \cdot h \\ h \end{bmatrix} &\leq b.\end{aligned}$$

- (c) Assume that at time $t = 0$ the system is at a given state, and what we want is to have it at time N in another given state. Due to linearity, we lose nothing when assuming that the initial state is 0, and the target state is a given vector x_* . The design specifications are as follows:

- If there are no disturbances (i.e., the initial state is exactly 0, and all d 's are zeros, we want the trajectory (let us call it the *nominal one*) $w^N = w_{\text{nom}}^N$ to have x_N exactly equal to x_* . In other words, we say that the nominal range of $[z; d^N]$ is just the origin, and when $[z; d^N]$ is in this range, the state at time N should be x_* .

- When $[z; d^N]$ deviates from 0, the trajectory w^N will deviate from the nominal trajectory w_{nom}^N , and we want the scaled uniform norm $\|\text{Diag}\{\beta\}(w^N - w_{\text{nom}}^N)\|_\infty$ not to exceed a multiple, with the smallest possible factor, of the scaled uniform norm $\|\text{Diag}\{\gamma\}[z; d^N]\|_\infty$ of the deviation of $[z; d^n]$ from its normal range (the origin). Here β and γ are positive weight vectors of appropriate sizes.

Prove that the outlined design specifications reduce to the following LP in variables $\vec{\eta}^N = (\vec{h}, \mathbf{H})$ and additional matrix variable Λ of appropriate size:

$$\min_{\substack{\vec{\eta}^N = (\vec{h}, \mathbf{H}), \\ \Lambda, \tau, \alpha}} \left\{ \alpha : \begin{array}{l} (a) \quad \mathbf{H} \text{ is block lower triangular} \\ (b) \quad \text{the last } n_x \text{ entries in } u2x \cdot \vec{h} \text{ form the vector } x_* \\ (c) \quad \Lambda_{ij} \geq 0 \forall i, j \\ (d) \quad \Lambda \begin{bmatrix} \text{Diag}\{\gamma\} \\ -\text{Diag}\{\gamma\} \end{bmatrix} = \begin{bmatrix} u2x \cdot \mathbf{H} \cdot \zeta v + \zeta 2x \\ -u2x \cdot \mathbf{H} \cdot \zeta v - \zeta 2x \end{bmatrix} \\ (e) \quad \Lambda \cdot \mathbf{1} \leq \alpha \begin{bmatrix} \beta \\ \beta \end{bmatrix} \end{array} \right\} \quad (1.3.27)$$

where $\mathbf{1}$ is the all-one vector of appropriate dimension.

3. Task 2: Implement the approach summarized in (1.3.27) in the following situation:

Boeing 747 is flying horizontally along the X -axis in the XZ plane with the velocity 774 ft/sec at the altitude 40000 ft and has 100 sec to change the altitude to 41000 ft. You should plan this maneuver.

The details are as follows¹⁴. First, you should keep in mind that in the description to follow the state variables are *deviations* of the “physical” state variables (like velocity, altitude, etc.) from their values at the “physical” state of the aircraft at the initial time instant, and not the physical states themselves, and similarly for the controls – they are deviations of actual controls from some “background” controls. The linear dynamics approximation is enough accurate, provided that these deviations are not too large. This linear dynamics is as follows:

- *State variables:* Δh : deviation in altitude, ft; Δu : deviation in velocity along the aircraft’s X -axis; Δv : deviation in velocity orthogonal to the aircraft’s axis, ft/sec (positive is down); $\Delta\theta$: deviation in angle θ between the aircraft’s axis and the X -axis (positive is up), crad (hundredths of radian); q : angular velocity of the aircraft (pitch rate), crad/sec.
 - *Disturbances:* u_w : wind velocity along the aircraft’s axis, ft/sec; v_w : wind velocity orthogonal to the aircraft’s axis, ft/sec. We assume that there is no side wind.
- Assuming that δ stays small, we will not distinguish between Δu (Δv) and the deviations of the aircraft’s velocities along the X -, respectively, Z -axes, and similarly for u_w and v_w .
- *Controls:* δ_e : elevator angle deviation (positive is down); δ_t : thrust deviation.
 - *Outputs:* velocity deviation Δu and climb rate $\dot{h} = -\Delta v + 7.74\theta$.

¹⁴Up to notation, the data below are taken from Lecture 14, of Prof. Stephen Boyd’s EE263 Lecture Notes, see <http://www.stanford.edu/~boyd/ee263/lectures/aircraft.pdf>

- *Linearized dynamics:*

$$\begin{aligned} \frac{d}{ds} \begin{bmatrix} \Delta h \\ \Delta u \\ \Delta v \\ q \\ \Delta \theta \end{bmatrix} &= \underbrace{\begin{bmatrix} 0 & 0 & -1 & 0 & 0 \\ 0 & -.003 & .039 & 0 & -.322 \\ 0 & -.065 & -.319 & 7.74 & 0 \\ 0 & .020 & -.101 & -.429 & 0 \\ 0 & 0 & 0 & 1 & 0 \end{bmatrix}}_{\mathcal{A}} \cdot \underbrace{\begin{bmatrix} \Delta h \\ \Delta u \\ \Delta v \\ q \\ \Delta \theta \end{bmatrix}}_{\chi(s)} + \underbrace{\begin{bmatrix} 0 & 0 \\ .01 & 1 \\ -.18 & -.04 \\ -1.16 & .598 \\ 0 & 0 \end{bmatrix}}_{\mathcal{B}} \underbrace{\begin{bmatrix} \delta_e \\ \delta_t \end{bmatrix}}_{v(s)} \\ &\quad + \underbrace{\begin{bmatrix} 0 & -1 \\ .003 & -.039 \\ .065 & .319 \\ -.020 & .101 \\ 0 & 0 \end{bmatrix}}_{\mathcal{R}} \cdot \underbrace{\begin{bmatrix} u_w \\ v_w \end{bmatrix}}_{\delta(s)} \\ y &= \underbrace{\begin{bmatrix} 0 & 1 & 0 & 0 & 0 \\ 0 & 0 & -1 & 0 & 7.74 \end{bmatrix}}_c \cdot \begin{bmatrix} \Delta h \\ \Delta u \\ \Delta v \\ q \\ \Delta \theta \end{bmatrix} \end{aligned}$$

- *From continuous to discrete time:* Note that our system is a *continuous time* one; we denote this time by s and measure it in seconds (this is why in the Boeing state dynamics the derivative with respect to time is denoted by $\frac{d}{ds}$, Boeing's state in a (continuous time) instant s is denoted by $\chi(s)$, etc.) Our current goal is to approximate this system by a discrete time one, in order to use our machinery for controller design. To this end, let us choose a “time resolution” $\Delta s > 0$, say, $\Delta s = 10$ sec, and look at the physical system at continuous time instants $t\Delta s$, $t = 0, 1, \dots$, so that the state x_t of the system at discrete time instant t is $\chi(t\Delta s)$, and similarly for controls, outputs, disturbances, etc. With this approach, 100 sec-long maneuver corresponds to the discrete time horizon $N = 10$.

We now should decide how to approximate the actual – continuous time – system with a discrete time one. The simplest way to do so would be just to replace differential equations with their finite-difference approximations:

$$x_{t+1} = \chi(t\Delta s + \Delta s) \approx \chi(t\Delta s) + (\Delta s \mathcal{A}) \chi(t\Delta s) + (\Delta s \mathcal{B}) v(t\Delta s) + (\Delta s \mathcal{R}) \delta(t\Delta s)$$

We can, however, do better than this. Let us associate with discrete time instant t the continuous time interval $[t\Delta s, (t+1)\Delta s]$, and assume (this is a quite reasonable assumption) that the continuous time controls are kept constant on these continuous time intervals:

$$t\Delta s \leq s < (t+1)\Delta s \Rightarrow v(s) \equiv u_t; \quad (*)$$

u_t will be our discrete time controls. Let us now define the matrices of the discrete time dynamics in such a way that with our piecewise constant continuous time controls, the *discrete time states* x_t of the discrete time system will be exactly the same as the states $\chi(t\Delta s)$ of the continuous time system. To this end let us note that with the piecewise constant continuous time controls (*), we have exact equations

$$\begin{aligned} \chi((t+1)\Delta s) &= [\exp\{\Delta s \mathcal{A}\}] \chi(t\Delta s) + \left[\int_0^{\Delta s} \exp\{r \mathcal{A}\} \mathcal{B} dr \right] u_t \\ &\quad + \left[\int_0^{\Delta s} \exp\{r \mathcal{A}\} \mathcal{R} \delta((t+1)\Delta s - r) dr \right], \end{aligned}$$

where the exponent of a (square) matrix can be defined by either one of the straightforward matrix analogies of the (equivalent to each other) definitions of the univariate exponent:

$$\begin{aligned}\exp\{B\} &= \lim_{n \rightarrow \infty} (I + n^{-1}B)^n, \text{ or} \\ \exp\{B\} &= \sum_{n=0}^{\infty} \frac{1}{n!} B^n.\end{aligned}$$

We see that when setting

$$A = \exp\{\Delta s \mathcal{A}\}, B = \left[\int_0^{\Delta s} \exp\{r \mathcal{A}\} \mathcal{B} dr \right], e_t = \int_0^{\Delta s} \exp\{r \mathcal{A}\} \mathcal{R} \delta((t+1)\Delta s - r) dr,$$

we get exact equalities

$$x_{t+1} := \chi(t\Delta s) = Ax_t + Bu_t + e_t, x_0 = \chi(0) \quad (\$)$$

If we now assume (this again makes complete sense) that continuous time outputs are measured at continuous time instants $t\Delta s$, $t = 0, 1, \dots$, we can augment the above discrete time dynamics with description of discrete time outputs:

$$y_t = \mathcal{C}x_t.$$

Now imagine that the decisions on u_t are made as follows: at “continuous time” instant $t\Delta s$, after y_t is measured, we immediately specify u_t as an affine function of y_1, \dots, y_t and keep the controls in the “physical time” interval $[t\Delta s, (t+1)\Delta s]$ at the level u_t . Now both the dynamics and the candidate affine controllers in continuous time fit their discrete time counterparts, and we can pass from affine output based controllers to affine purified output based ones, and utilize the outlined approach for synthesis of a.p.o.b. controllers to build efficiently a controller for the “true” continuous time system. Note that with this approach we should not bother much on how small is Δs , since our discrete time approximation reproduces exactly the behaviour of the continuous time system along the discrete grid $t\Delta s$, $t = 0, 1, \dots$. The only restriction on Δs comes from the desire to make design specifications expressed in terms of discrete time trajectory meaningful in terms of the actual continuous time behavior, which usually is not too difficult.

The only difficulty which still remains unresolved is how to translate our a priori information on continuous time disturbances $\delta(s)$ into information on their discrete time counterparts e_t . For example, when assuming that the normal range of the continuous time disturbances $\delta(\cdot)$ is given by an upper bound $\forall s, \|\delta(s)\| \leq \rho$ on a given norm $\|\cdot\|$ of the disturbance, the “translated” information on e_t is $e_t \in \rho E$, where $E = \left\{ e = \int_0^{\Delta s} \exp\{r \mathcal{A}\} \mathcal{R} \delta(r) dr : \|\delta(r)\| \leq 1, 0 \leq r \leq \Delta s \right\}$.¹⁵ We then can set $d_t = e_t$, $R = I_{n_x}$ and say that the normal range of the discrete time disturbances d_t is $d_t \in E$ for all t . A disadvantage of this approach is that for our purposes, we need E to admit a simple representation, which not necessary is the case with E “as it is.” Here is a (slightly conservative) way to resolve this difficulty. Assume that

¹⁵Note that this “translation” usually makes e_t vectors of the same dimension as the state vectors x_t , even when the dimension of the “physical disturbances” $\delta(t)$ is less than the dimension of the state vectors. This is natural: continuous time dynamics usually “spreads” the potential contribution, accumulated over a continuous time interval $[t\Delta s, (t+1)\Delta s]$, of low-dimensional time-varying disturbances $\delta(s)$ to the state $\chi((t+1)\Delta s)$ over a full-dimensional domain in the state space.

the normal range of the continuous time disturbance is given as $\|\delta(s)\|_\infty \leq \rho \forall s$. Let $\text{abs}[M]$ be the matrix comprised of the moduli of the entries of a matrix M . When $\|\delta(s)\|_\infty \leq 1$ for all s , we clearly have

$$\text{abs} \left[\int_0^{\Delta s} \exp\{r\mathcal{A}\} \mathcal{R} \delta(r) dr \right] \leq r := \int_0^{\Delta s} \text{abs}[\exp\{r\mathcal{A}\} \mathcal{R}] \cdot \mathbf{1} dr,$$

meaning that $E \subset \{\text{Diag}\{r\}d : \|d\|_\infty \leq 1\}$. In other words, we can only increase the set of allowed disturbances e_t by assuming that they are of the form $e_t = Rd_t$ with $R = \text{Diag}\{r\}$ and $\|d_t\|_\infty \leq \rho$. As a result, every controller for the discrete time system

$$\begin{aligned} x_{t+1} &= Ax_t + Bu_t + Rd_t, \\ y_t &= Cx_t \end{aligned}$$

with just defined A, B, R and $C = \mathcal{C}$ which meets desired design specifications, the normal range of d^N being $\{\|d_t\|_\infty \leq \rho \forall t\}$, definitely meets the design specifications for the system (\$\\$), the normal range of the disturbances e_t being $e_t \in \rho E$ for all t .

Now you are prepared to carry out your task:

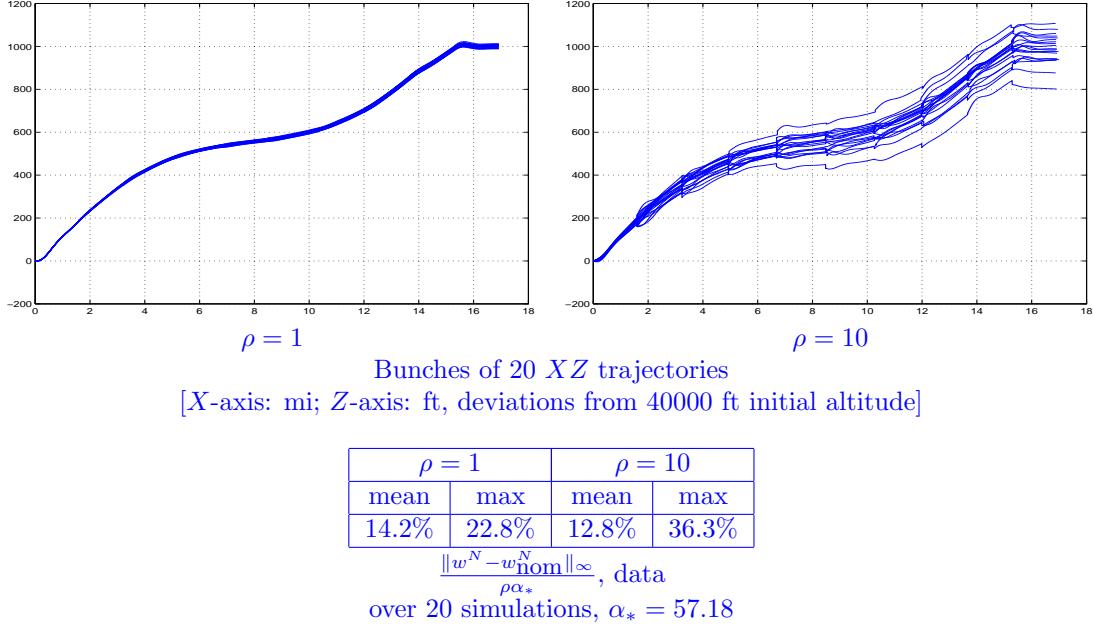
- (a) Implement the outlined strategy to the Boeing maneuver example. The still missing components of the setup are as follows:

$$\begin{aligned} \text{time resolution: } \Delta s &= 10\text{sec} \Rightarrow N = 10; \\ \text{state at } s = 0: \quad x_0 &= [\Delta h = 0; \Delta u = 0; \Delta v = 0; q = 0; \Delta \delta = 0] \\ &\quad [\text{cruiser horizontal flight at the speed } 774+73.3 \text{ ft/sec, altitude } 40000 \text{ ft}] \\ \text{state at } s = 100: \quad x_{10} &= [\Delta h = 1000; \Delta u = 0; \Delta v = 0; q = 0; \Delta \delta = 0] \\ &\quad [\text{cruiser horizontal flight at the speed } 774+73.3 \text{ ft/sec, altitude } 41000 \text{ ft}] \\ \text{weights in (1.3.27): } \gamma &= \mathbf{1}, \beta = \mathbf{1} \end{aligned}$$

($774 \text{ ft/sec} \approx 528 \text{ mph}$ is the aircraft speed w.r.t to the air, $73.3 \text{ ft/sec} = 50 \text{ mph}$ is the steady-state tail wind).

- Specify the a.o.p.b. control as given by an optimal solution to (1.3.27). What is the optimal value α_* in the problem?
- With the resulting control law, specify the nominal state-control trajectories of the discrete and the continuous time models of the maneuver (i.e., zero initial state, no wind); plot the corresponding XZ trajectory of the aircraft.
- Run 20 simulations of the discrete time and the continuous time models of the maneuver when the initial state and the disturbances are randomly selected vectors of uniform norm not exceeding $\rho = 1$ (which roughly corresponds to up to 1 mph randomly oriented in the XZ plane deviations of the actual wind velocity from the steady-state tail 50 mph one) and plot the resulting bunch of the aircraft's XZ trajectories. Check that the uniform deviation of the state-control trajectories from the nominal one all the time does not exceed $\alpha_*\rho$ (as it should be due to the origin of α_*). What is the mean, over the observed trajectories, ratio of the deviation to $\alpha\rho$?
- Repeat the latter experiment with ρ increased to 10 (up to 10 mph deviations in the wind velocity).

My results are as follows:



1.4 From Linear to Conic Programming

Linear Programming models cover numerous applications. Whenever applicable, LP allows to obtain useful quantitative and qualitative information on the problem at hand. The specific analytic structure of LP programs gives rise to a number of general results (e.g., those of the LP Duality Theory) which provide us in many cases with valuable insight and understanding. At the same time, this analytic structure underlies some specific computational techniques for LP; these techniques, which by now are perfectly well developed, allow to solve routinely quite large (tens/hundreds of thousands of variables and constraints) LP programs. Nevertheless, there are situations in reality which cannot be covered by LP models. To handle these “essentially nonlinear” cases, one needs to extend the basic theoretical results and computational techniques known for LP beyond the bounds of Linear Programming.

For the time being, the widest class of optimization problems to which the basic results of LP were extended, is the class of *convex* optimization programs. There are several equivalent ways to define a general convex optimization problem; the one we are about to use is not the traditional one, but it is well suited to encompass the range of applications we intend to cover in our course.

When passing from a generic LP problem

$$\min_x \left\{ c^T x \mid Ax \geq b \right\} \quad [A : m \times n] \quad (\text{LP})$$

to its nonlinear extensions, we should expect to encounter some nonlinear components in the problem. The traditional way here is to say: “Well, in (LP) there are a linear objective function $f(x) = c^T x$ and inequality constraints $f_i(x) \geq b_i$ with linear functions $f_i(x) = a_i^T x$, $i = 1, \dots, m$. Let us allow some/all of these functions f, f_1, \dots, f_m to be nonlinear.” In contrast to this traditional way, we intend to keep the objective and the constraints linear, but introduce “nonlinearity” in the inequality sign \geq .

1.4.1 Orderings of \mathbf{R}^m and cones

The constraint inequality $Ax \geq b$ in (LP) is an inequality between vectors; as such, it requires a definition, and the definition is well-known: given two vectors $a, b \in \mathbf{R}^m$, we write $a \geq b$, if the coordinates of a majorate the corresponding coordinates of b :

$$a \geq b \Leftrightarrow \{a_i \geq b_i, i = 1, \dots, m\}. \quad (" \geq ")$$

In the latter relation, we again meet with the inequality sign \geq , but now it stands for the “arithmetic \geq ” – a well-known relation between real numbers. The above “coordinate-wise” partial ordering of vectors in \mathbf{R}^m satisfies a number of basic properties of the standard ordering of reals; namely, for all vectors $a, b, c, d, \dots \in \mathbf{R}^m$ one has

1. *Reflexivity*: $a \geq a$;
2. *Anti-symmetry*: if both $a \geq b$ and $b \geq a$, then $a = b$;
3. *Transitivity*: if both $a \geq b$ and $b \geq c$, then $a \geq c$;
4. *Compatibility with linear operations*:
 - (a) *Homogeneity*: if $a \geq b$ and λ is a nonnegative real, then $\lambda a \geq \lambda b$
("One can multiply both sides of an inequality by a nonnegative real")
 - (b) *Additivity*: if both $a \geq b$ and $c \geq d$, then $a + c \geq b + d$
("One can add two inequalities of the same sign").

It turns out that

- A significant part of the nice features of LP programs comes from the fact that the vector inequality \geq in the constraint of (LP) satisfies the properties 1. – 4.;
- The standard inequality “ \geq ” is neither the only possible, nor the only interesting way to define the notion of a vector inequality fitting the axioms 1. – 4.

As a result,

A generic optimization problem which looks exactly the same as (LP), up to the fact that the inequality \geq in (LP) is now replaced with an ordering which differs from the component-wise one, inherits a significant part of the properties of LP problems. Specifying properly the ordering of vectors, one can obtain from (LP) generic optimization problems covering many important applications which cannot be treated by the standard LP.

To the moment what is said is just a declaration. Let us look how this declaration comes to life.

We start with clarifying the “geometry” of a “vector inequality” satisfying the axioms 1. – 4. Thus, we consider vectors from a finite-dimensional Euclidean space \mathbf{E} with an inner product $\langle \cdot, \cdot \rangle$ and assume that \mathbf{E} is equipped with a partial ordering (called also vector inequality), let it be denoted by \succeq : in other words, we say what are the pairs of vectors a, b from \mathbf{E} linked by the inequality $a \succeq b$. We call the ordering “good”, if it obeys the axioms 1. – 4., and are interested to understand what are these good orderings.

Our first observation is:

A. A good vector inequality \succeq is completely identified by the set \mathbf{K} of \succeq -nonnegative vectors:

$$\mathbf{K} = \{a \in \mathbf{E} : a \succeq 0\}.$$

Namely,

$$a \succeq b \Leftrightarrow a - b \succeq 0 \quad [\Leftrightarrow a - b \in \mathbf{K}].$$

Indeed, let $a \succeq b$. By 1. we have $-b \succeq -b$, and by 4.(b) we may add the latter inequality to the former one to get $a - b \succeq 0$. Vice versa, if $a - b \succeq 0$, then, adding to this inequality the one $b \succeq b$, we get $a \succeq b$.

The set \mathbf{K} in Observation A cannot be arbitrary. It is easy to verify that it must be a *pointed cone*, i.e., it must satisfy the following conditions:

1. \mathbf{K} is nonempty and closed under addition:

$$a, a' \in \mathbf{K} \Rightarrow a + a' \in \mathbf{K};$$

2. \mathbf{K} is a conic set:

$$a \in \mathbf{K}, \lambda \geq 0 \Rightarrow \lambda a \in \mathbf{K}.$$

3. \mathbf{K} is pointed:

$$a \in \mathbf{K} \text{ and } -a \in \mathbf{K} \Rightarrow a = 0.$$

Geometrically: \mathbf{K} does not contain straight lines passing through the origin.

Exercise 1.3 Prove that the outlined properties of \mathbf{K} are necessary and sufficient for the vector inequality $a \succeq b \Leftrightarrow a - b \in \mathbf{K}$ to be good.

Thus, every pointed cone \mathbf{K} in \mathbf{E} induces a partial ordering on \mathbf{E} which satisfies the axioms 1. – 4. We denote this ordering by $\geq_{\mathbf{K}}$:

$$a \geq_{\mathbf{K}} b \Leftrightarrow a - b \geq_{\mathbf{K}} 0 \Leftrightarrow a - b \in \mathbf{K}.$$

What is the cone responsible for the standard coordinate-wise ordering \geq on $\mathbf{E} = \mathbf{R}^m$ we have started with? The answer is clear: this is the cone comprised of vectors with nonnegative entries – the *nonnegative orthant*

$$\mathbf{R}_+^m = \{x = (x_1, \dots, x_m)^T \in \mathbf{R}^m : x_i \geq 0, i = 1, \dots, m\}.$$

(Thus, in order to express the fact that a vector a is greater than or equal to, in the component-wise sense, to a vector b , we were supposed to write $a \geq_{\mathbf{R}_+^m} b$. However, we are not going to be that formal and shall use the standard shorthand notation $a \geq b$.)

The nonnegative orthant \mathbf{R}_+^m is not just a pointed cone; it possesses two useful additional properties:

I. The cone is closed: if a sequence of vectors a^i from the cone has a limit, the latter also belongs to the cone.

II. The cone possesses a nonempty interior: there exists a vector such that a ball of positive radius centered at the vector is contained in the cone.

These additional properties are very important. For example, \mathbf{I} is responsible for the possibility to pass to the term-wise limit in an inequality:

$$a^i \geq b^i \quad \forall i, \quad a^i \rightarrow a, b^i \rightarrow b \text{ as } i \rightarrow \infty \Rightarrow a \geq b.$$

It makes sense to restrict ourselves with good partial orderings coming from cones \mathbf{K} sharing the properties **I**, **II**. Thus,

From now on, speaking about vector inequalities $\geq_{\mathbf{K}}$, we always assume that the underlying set \mathbf{K} is a pointed and closed cone with a nonempty interior.

Note that the closedness of \mathbf{K} makes it possible to pass to limits in $\geq_{\mathbf{K}}$ -inequalities:

$$a^i \geq_{\mathbf{K}} b^i, \quad a^i \rightarrow a, b^i \rightarrow b \text{ as } i \rightarrow \infty \Rightarrow a \geq_{\mathbf{K}} b.$$

The nonemptiness of the interior of \mathbf{K} allows to define, along with the “non-strict” inequality $a \geq_{\mathbf{K}} b$, also the strict inequality according to the rule

$$a >_{\mathbf{K}} b \Leftrightarrow a - b \in \text{int } \mathbf{K},$$

where $\text{int } K$ is the interior of the cone \mathbf{K} . E.g., the strict coordinate-wise inequality $a >_{\mathbf{R}_+^m} b$ (shorthand: $a > b$) simply says that the coordinates of a are strictly greater, in the usual arithmetic sense, than the corresponding coordinates of b .

Examples. The partial orderings we are especially interested in are given by the following cones:

- The nonnegative orthant \mathbf{R}_+^m in \mathbf{R}^n ;
- The *Lorentz* (or the *second-order*, or the less scientific name the *ice-cream*) cone

$$\mathbf{L}^m = \left\{ x = (x_1, \dots, x_{m-1}, x_m)^T \in \mathbf{R}^m : x_m \geq \sqrt{\sum_{i=1}^{m-1} x_i^2} \right\}$$

- The *semidefinite cone* \mathbf{S}_+^m . This cone “lives” in the space $\mathbf{E} = \mathbf{S}^m$ of $m \times m$ symmetric matrices (equipped with the Frobenius inner product $\langle A, B \rangle = \text{Tr}(AB) = \sum_{i,j} A_{ij}B_{ij}$) and consists of all $m \times m$ matrices A which are *positive semidefinite*, i.e.,

$$A = A^T; \quad x^T Ax \geq 0 \quad \forall x \in \mathbf{R}^m.$$

1.4.2 “Conic programming” – what is it?

Let \mathbf{K} be a *regular* cone in \mathbf{E} , regularity meaning that the cone is convex, pointed, closed and with a nonempty interior. Given an *objective* $c \in \mathbf{R}^n$, a linear mapping $x \mapsto Ax : \mathbf{R}^n \rightarrow \mathbf{E}$ and a *right hand side* $b \in \mathbf{E}$, consider the optimization problem

$$\min_x \left\{ c^T x \mid Ax \geq_{\mathbf{K}} b \right\} \tag{CP}.$$

We shall refer to (CP) as to a *conic problem* associated with the cone \mathbf{K} . Note that the only difference between this program and an LP problem is that the latter deals with the particular choice $\mathbf{E} = \mathbf{R}^m$, $\mathbf{K} = \mathbf{R}_+^m$. With the formulation (CP), we get a possibility to cover a much wider spectrum of applications which cannot be captured by LP; we shall look at numerous examples in the sequel.

1.4.3 Conic Duality

Aside of algorithmic issues, the most important theoretical result in Linear Programming is the LP Duality Theorem; can this theorem be extended to conic problems? What is the extension?

The source of the LP Duality Theorem was the desire to get in a systematic way a lower bound on the optimal value c^* in an LP program

$$c^* = \min_x \left\{ c^T x \mid Ax \geq b \right\}. \quad (\text{LP})$$

The bound was obtained by looking at the inequalities of the type

$$\langle \lambda, Ax \rangle \equiv \lambda^T Ax \geq \lambda^T b \quad (\text{Cons}(\lambda))$$

with weight vectors $\lambda \geq 0$. By its origin, an inequality of this type is a consequence of the system of constraints $Ax \geq b$ of (LP), i.e., it is satisfied at every solution to the system. Consequently, whenever we are lucky to get, as the left hand side of (Cons(λ)), the expression $c^T x$, i.e., whenever a nonnegative weight vector λ satisfies the relation

$$A^T \lambda = c,$$

the inequality (Cons(λ)) yields a lower bound $b^T \lambda$ on the optimal value in (LP). And the dual problem

$$\max \left\{ b^T \lambda : \lambda \geq 0, A^T \lambda = c \right\}$$

was nothing but the problem of finding the best lower bound one can get in this fashion.

The same scheme can be used to develop the dual to a conic problem

$$\min \left\{ c^T x : Ax \geq_{\mathbf{K}} b \right\}, \quad \mathbf{K} \subset \mathbf{E}. \quad (\text{CP})$$

Here the only step which needs clarification is the following one:

(?) What are the “admissible” weight vectors λ , i.e., the vectors such that the scalar inequality

$$\langle \lambda, Ax \rangle \geq \langle \lambda, b \rangle$$

is a consequence of the vector inequality $Ax \geq_{\mathbf{K}} b$?

In the particular case of coordinate-wise partial ordering, i.e., in the case of $\mathbf{E} = \mathbf{R}^m$, $\mathbf{K} = \mathbf{R}_+^m$, the admissible vectors were those with nonnegative coordinates. These vectors, however, not necessarily are admissible for an ordering $\geq_{\mathbf{K}}$ when \mathbf{K} is different from the nonnegative orthant:

Example 1.4.1 Consider the ordering $\geq_{\mathbf{L}^3}$ on $\mathbf{E} = \mathbf{R}^3$ given by the 3-dimensional ice-cream cone:

$$\begin{pmatrix} a_1 \\ a_2 \\ a_3 \end{pmatrix} \geq_{\mathbf{L}^3} \begin{pmatrix} 0 \\ 0 \\ 0 \end{pmatrix} \Leftrightarrow a_3 \geq \sqrt{a_1^2 + a_2^2}.$$

The inequality

$$\begin{pmatrix} -1 \\ -1 \\ 2 \end{pmatrix} \geq_{\mathbf{L}^3} \begin{pmatrix} 0 \\ 0 \\ 0 \end{pmatrix}$$

is valid; however, aggregating this inequality with the aid of a positive weight vector $\lambda = \begin{pmatrix} 1 \\ 1 \\ 0.1 \end{pmatrix}$, we get the false inequality

$$-1.8 \geq 0.$$

Thus, not every nonnegative weight vector is admissible for the partial ordering $\geq_{\mathbf{L}^3}$.

To answer the question (?) is the same as to say what are the weight vectors λ such that

$$\forall a \geq_{\mathbf{K}} 0 : \langle \lambda, a \rangle \geq 0. \quad (1.4.1)$$

Whenever λ possesses the property (1.4.1), the scalar inequality

$$\langle \lambda, a \rangle \geq \langle \lambda, b \rangle$$

is a consequence of the vector inequality $a \geq_{\mathbf{K}} b$:

$$\begin{aligned} & a \geq_{\mathbf{K}} b \\ \Leftrightarrow & a - b \geq_{\mathbf{K}} 0 \quad [\text{additivity of } \geq_{\mathbf{K}}] \\ \Rightarrow & \langle \lambda, a - b \rangle \geq 0 \quad [\text{by (1.4.1)}] \\ \Leftrightarrow & \langle \lambda, a \rangle \geq \lambda^T b. \quad \blacksquare \end{aligned}$$

Vice versa, if λ is an admissible weight vector for the partial ordering $\geq_{\mathbf{K}}$:

$$\forall (a, b : a \geq_{\mathbf{K}} b) : \langle \lambda, a \rangle \geq \langle \lambda, b \rangle$$

then, of course, λ satisfies (1.4.1).

Thus the weight vectors λ which are admissible for a partial ordering $\geq_{\mathbf{K}}$ are exactly the vectors satisfying (1.4.1), or, which is the same, the vectors from the set

$$\mathbf{K}_* = \{\lambda \in \mathbf{E} : \langle \lambda, a \rangle \geq 0 \quad \forall a \in \mathbf{K}\}.$$

The set \mathbf{K}_* is comprised of vectors whose inner products with *all* vectors from \mathbf{K} are nonnegative. \mathbf{K}_* is called the *cone dual to \mathbf{K}* . The name is legitimate due to the following fact (see Section B.2.7.B):

Theorem 1.4.1 [Properties of the dual cone] *Let \mathbf{E} be a finite-dimensional Euclidean space with inner product $\langle \cdot, \cdot \rangle$ and let $K \subset \mathbf{E}$ be a nonempty set. Then*

(i) *The set*

$$K_* = \{\lambda \in \mathbf{E}^m : \langle \lambda, a \rangle \geq 0 \quad \forall a \in K\}$$

is a closed cone.

(ii) *If $\text{int } K \neq \emptyset$, then K_* is pointed.*

(iii) *If K is a closed convex pointed cone, then $\text{int } K_* \neq \emptyset$.*

(iv) *If K is a closed cone, then so is K_* , and the cone dual to K_* is K itself:*

$$(K_*)_* = K.$$

An immediate corollary of the Theorem is as follows:

Corollary 1.4.1 *A set $K \subset \mathbf{E}$ is a regular (i.e., closed, convex, pointed and with a nonempty interior) cone if and only if the set K_* is so.*

From the dual cone to the problem dual to (CP). Now we are ready to derive the dual problem of a conic problem (CP). As in the case of Linear Programming, we start with the observation that whenever x is a feasible solution to (CP) and λ is an admissible weight vector, i.e., $\lambda \in \mathbf{K}_*$, then x satisfies the scalar inequality

$$(A^* \lambda)^T x \equiv \langle \lambda, Ax \rangle \geq \langle \lambda, b \rangle \quad ^{16)}$$

– this observation is an immediate consequence of the definition of \mathbf{K}_* . It follows that whenever λ_* is an admissible weight vector satisfying the relation

$$A^* \lambda = c,$$

one has

$$c^T x = (A^* \lambda)^T x = \langle \lambda, Ax \rangle \geq \langle b, \lambda \rangle$$

for all x feasible for (CP), so that the quantity $\langle b, \lambda \rangle$ is a lower bound on the optimal value of (CP). The best bound one can get in this fashion is the optimal value in the problem

$$\max \{ \langle b, \lambda \rangle : A^* \lambda = c, \lambda \geq_{\mathbf{K}_*} 0 \} \quad (\text{CD})$$

and this program is called the program *dual to (CP)*.

So far, what we know about the duality we have just introduced is the following

Proposition 1.4.1 [Weak Conic Duality Theorem] *The optimal value of (CD) is a lower bound on the optimal value of (CP).*

1.4.4 Geometry of the primal and the dual problems

The structure of problem (CD) looks quite different from the one of (CP). However, a more careful analysis demonstrates that the difference in structures comes just from how we represent the data: geometrically, the problems are completely similar. Indeed, in (CD) we are asked to maximize a linear objective $\langle b, \lambda \rangle$ over the intersection of an affine plane $L_* = \{\lambda : A^* \lambda = c\}$ with the cone \mathbf{K}_* . And what about (CP)? Let us pass in this problem from the “true design variables” x to their images $y = Ax - b \in \mathbf{E}$. When x runs through \mathbf{R}^n , y runs through the affine plane $L = \{y = Ax - b : x \in \mathbf{R}^n\}$; $x \in \mathbf{R}^n$ is feasible for (CP) if and only if the corresponding $y = Ax - b$ belongs to the cone \mathbf{K} . Thus, in (CP) we also deal with the intersection of an affine plane, namely, L , and a cone, namely, \mathbf{K} . Now assume that our objective $c^T x$ can be expressed in terms of $y = Ax - b$:

$$c^T x = \langle d, Ax - b \rangle + \text{const.}$$

This assumption is clearly equivalent to the inclusion

$$c \in \text{Im } A^*. \quad (1.4.2)$$

¹⁶⁾ For a linear operator $x \mapsto Ax : \mathbf{R}^n \rightarrow \mathbf{E}$, A^* is the conjugate operator given by the identity

$$\langle y, Ax \rangle = x^T A y \quad \forall (y \in \mathbf{E}, x \in \mathbf{R}^n).$$

When representing the operators by their matrices in *orthogonal* bases in the argument and the range spaces, the matrix representing the conjugate operator is exactly the transpose of the matrix representing the operator itself.

Indeed, in the latter case we have $c = A^*d$ for some d , whence

$$c^T x = \langle A^*d, x \rangle = \langle d, Ax \rangle = \langle d, Ax - b \rangle + \langle d, b \rangle \quad \forall x. \quad (1.4.3)$$

In the case of (1.4.2) the primal problem (CP) can be posed equivalently as the following problem:

$$\min_y \{ \langle d, y \rangle : y \in L, y \geq_{\mathbf{K}} 0 \},$$

where $L = \text{Im}A - b$ and d is (any) vector satisfying the relation $A^*d = c$. Thus,

In the case of (1.4.2) the primal problem, geometrically, is the problem of minimizing a linear form over the intersection of the affine plane L with the cone \mathbf{K} , and the dual problem, similarly, is to maximize another linear form over the intersection of the affine plane L_ with the dual cone \mathbf{K}_* .*

Now, what happens if the condition (1.4.2) is not satisfied? The answer is very simple: in this case (CP) makes no sense – it is either unbounded below, or infeasible.

Indeed, assume that (1.4.2) is not satisfied. Then, by Linear Algebra, the vector c is not orthogonal to the null space of A , so that there exists e such that $Ae = 0$ and $c^T e > 0$. Now let x be a feasible solution of (CP); note that all points $x - \mu e$, $\mu \geq 0$, are feasible, and $c^T(x - \mu e) \rightarrow \infty$ as $\mu \rightarrow \infty$. Thus, when (1.4.2) is not satisfied, problem (CP), whenever feasible, is unbounded below.

From the above observation we see that if (1.4.2) is not satisfied, then we may reject (CP) from the very beginning. Thus, from now on we assume that (1.4.2) is satisfied. In fact in what follows we make a bit stronger assumption:

A. *The mapping A is of full column rank, i.e., it has trivial null space.*

Assuming that the mapping $x \mapsto Ax$ has the trivial null space (“we have eliminated from the very beginning the redundant degrees of freedom – those not affecting the value of Ax ”), the equation

$$A^*d = q$$

is solvable for every right hand side vector q .

In view of **A**, problem (CP) can be reformulated as a problem (P) of minimizing a linear objective $\langle d, y \rangle$ over the intersection of an affine plane L and a cone \mathbf{K} . Conversely, a problem (P) of this latter type can be posed in the form of (CP) – to this end it suffices to represent the plane L as the image of an affine mapping $x \mapsto Ax - b$ (i.e., to parameterize somehow the feasible plane) and to “translate” the objective $\langle d, y \rangle$ to the space of x -variables – to set $c = A^*d$, which yields

$$y = Ax - b \Rightarrow \langle d, y \rangle = c^T x + \text{const.}$$

Thus, when dealing with a conic problem, we may pass from its “analytic form” (CP) to the “geometric form” (P) and vice versa.

What are the relations between the “geometric data” of the primal and the dual problems? We already know that the cone \mathbf{K}_* associated with the dual problem is dual of the cone \mathbf{K} associated with the primal one. What about the feasible planes L and L_* ? The answer is

simple: they are orthogonal to each other! More exactly, the affine plane L is the translation, by vector $-b$, of the linear subspace

$$\mathcal{L} = \text{Im } A \equiv \{y = Ax : x \in \mathbf{R}^n\}.$$

And L_* is the translation, by any solution λ_0 of the system $A^*\lambda = c$, e.g., by the solution d to the system, of the linear subspace

$$\mathcal{L}_* = \text{Null}(A^*) \equiv \{\lambda : A^*\lambda = 0\}.$$

A well-known fact of Linear Algebra is that the linear subspaces \mathcal{L} and \mathcal{L}_* are orthogonal complements of each other:

$$\mathcal{L} = \{y : \langle y, \lambda \rangle = 0 \quad \forall \lambda \in \mathcal{L}_*\}; \quad \mathcal{L}_* = \{\lambda : \langle y, \lambda \rangle = 0 \quad \forall y \in \mathcal{L}\}.$$

Thus, we come to a nice geometrical conclusion:

A conic problem¹⁷⁾ (CP) is the problem

$$\min_y \{\langle d, y \rangle : y \in \mathcal{L} - b, \quad y \geq_{\mathbf{K}} 0\} \quad (\text{P})$$

of minimizing a linear objective $\langle d, y \rangle$ over the intersection of a cone \mathbf{K} with an affine plane $L = \mathcal{L} - b$ given as a translation, by vector $-b$, of a linear subspace \mathcal{L} .

The dual problem is the problem

$$\max_{\lambda} \{\langle b, \lambda \rangle : \lambda \in \mathcal{L}^\perp + d, \quad \lambda \geq_{\mathbf{K}*} 0\}. \quad (\text{CD})$$

of maximizing the linear objective $\langle b, \lambda \rangle$ over the intersection of the dual cone \mathbf{K}_ with an affine plane $L_* = \mathcal{L}^\perp + d$ given as a translation, by the vector d , of the orthogonal complement \mathcal{L}^\perp of \mathcal{L} .*

What we get is an extremely transparent geometric description of the primal-dual pair of conic problems (P), (CD). Note that the duality is completely symmetric: the problem dual to (CD) is (P)! Indeed, we know from Theorem 1.4.1 that $(\mathbf{K}_*)_* = \mathbf{K}$, and of course $(\mathcal{L}^\perp)^\perp = \mathcal{L}$. Switch from maximization to minimization corresponds to the fact that the “shifting vector” in (P) is $(-b)$, while the “shifting vector” in (CD) is d . The geometry of the primal-dual pair (P), (CD)

¹⁷⁾recall that we have restricted ourselves to the problems satisfying the assumption **A**

is illustrated on the below picture:

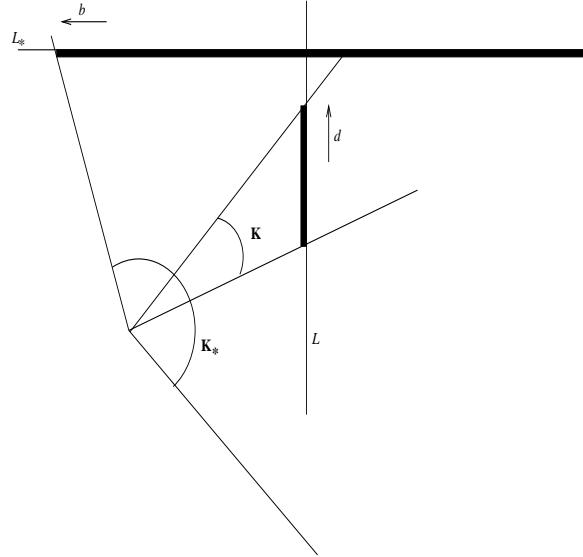


Figure 1.1. Primal-dual pair of conic problems
[bold: primal (vertical segment) and dual (horizontal ray) feasible sets]

Finally, note that in the case when (CP) is an LP program (i.e., in the case when \mathbf{K} is the nonnegative orthant), the “conic dual” problem (CD) is exactly the usual LP dual; this fact immediately follows from the observation that the cone dual to \mathbf{R}_+^m is \mathbf{R}_+^m itself.

We have explored the geometry of a primal-dual pair of conic problems: the “geometric data” of such a pair are given by a pair of dual to each other cones \mathbf{K}, \mathbf{K}_* in \mathbf{E} and a pair of affine planes $L = \mathcal{L} - b, L_* = \mathcal{L}^\perp + d$, where \mathcal{L} is a linear subspace in \mathbf{E} and \mathcal{L}^\perp is its orthogonal complement. The first problem from the pair – let it be called (P) – is to minimize $\langle b, y \rangle$ over $y \in \mathbf{K} \cap L$, and the second (CD) is to maximize $\langle d, \lambda \rangle$ over $\lambda \in \mathbf{K}_* \cap L_*$. Note that the “geometric data” $(\mathbf{K}, \mathbf{K}_*, L, L_*)$ of the pair do not specify completely the problems of the pair: given L, L_* , we can uniquely define \mathcal{L} , but not the shift vectors $(-b)$ and d : b is known up to shift by a vector from \mathcal{L} , and d is known up to shift by a vector from \mathcal{L}^\perp . However, this non-uniqueness is of absolutely no importance: replacing a chosen vector $d \in L_*$ by another vector $d' \in L_*$, we pass from (P) to a new problem (P') which is completely equivalent to (P): indeed, both (P) and (P') have the same feasible set, and *on the (common) feasible plane L of the problems their objectives $\langle d, y \rangle$ and $\langle d', y \rangle$ differ from each other by a constant*:

$$y \in L = \mathcal{L} - b, d - d' \in \mathcal{L}^\perp \Rightarrow \langle d - d', y + b \rangle = 0 \Rightarrow \langle d - d', y \rangle = \langle -(d - d'), b \rangle \quad \forall y \in L.$$

Similarly, shifting b along \mathcal{L} , we do modify the objective in (CD), but in a trivial way – *on the feasible plane L_* of the problem the new objective differs from the old one by a constant*.

1.4.5 Conic Duality Theorem

The Weak Duality (Proposition 1.4.1) we have established so far for conic problems is much weaker than the Linear Programming Duality Theorem. Is it possible to get results similar to those of the LP Duality Theorem in the general conic case as well? The answer is affirmative, *provided that the primal problem (CP) is strictly feasible*, i.e., that there exists x such that $Ax - b >_{\mathbf{K}} 0$, or, geometrically, $L \cap \text{int } \mathbf{K} \neq \emptyset$.

The advantage of the geometrical definition of strict feasibility is that it is independent of the particular way in which the feasible plane is defined; hence, with this definition it is clear what does it mean that the dual problem (CD) is strictly feasible.

Our main result is the following

Theorem 1.4.2 [Conic Duality Theorem] *Consider a conic problem*

$$c^* = \min_x \{c^T x : Ax \geq_{\mathbf{K}} b\} \quad (\text{CP})$$

along with its conic dual

$$b^* = \max \{\langle b, \lambda \rangle : A^* \lambda = c, \lambda \geq_{\mathbf{K}_*} 0\}. \quad (\text{CD})$$

1) *The duality is symmetric: the dual problem is conic, and the problem dual to dual is the primal.*

2) *The value of the dual objective at every dual feasible solution λ is \leq the value of the primal objective at every primal feasible solution x , so that the duality gap*

$$c^T x - \langle b, \lambda \rangle$$

is nonnegative at every “primal-dual feasible pair” (x, λ) .

3.a) *If the primal (CP) is bounded below and strictly feasible (i.e. $Ax >_{\mathbf{K}} b$ for some x), then the dual (CD) is solvable and the optimal values in the problems are equal to each other: $c^* = b^*$.*

3.b) *If the dual (CD) is bounded above and strictly feasible (i.e., exists $\lambda >_{\mathbf{K}_*} 0$ such that $A^* \lambda = c$), then the primal (CP) is solvable and $c^* = b^*$.*

4) *Assume that at least one of the problems (CP), (CD) is bounded and strictly feasible. Then a primal-dual feasible pair (x, λ) is a pair of optimal solutions to the respective problems*

4.a) *if and only if*

$$\langle b, \lambda \rangle = c^T x \quad [\text{zero duality gap}]$$

and

4.b) *if and only if*

$$\langle \lambda, Ax - b \rangle = 0 \quad [\text{complementary slackness}]$$

Proof. 1): The result was already obtained when discussing the geometry of the primal and the dual problems.

2): This is the Weak Conic Duality Theorem.

3): Assume that (CP) is strictly feasible and bounded below, and let c^* be the optimal value of the problem. We should prove that the dual is solvable with the same optimal value. Since we already know that the optimal value of the dual is $\leq c^*$ (see 2)), all we need is to point out a dual feasible solution λ_* with $b^T \lambda_* \geq c^*$.

Consider the convex set

$$M = \{y = Ax - b : x \in \mathbf{R}^n, c^T x \leq c^*\}.$$

Let us start with the case of $c \neq 0$. We claim that in this case

- (i) The set M is nonempty;
- (ii) the plane M does not intersect the interior K of the cone \mathbf{K} : $M \cap \text{int } \mathbf{K} = \emptyset$.

(i) is evident (why?). To verify (ii), assume, on contrary, that there exists a point \bar{x} , $c^T \bar{x} \leq c^*$, such that $\bar{y} \equiv A\bar{x} - b >_{\mathbf{K}} 0$. Then, of course, $Ax - b >_{\mathbf{K}} 0$ for all x close enough to \bar{x} , i.e., all points x in a small enough neighbourhood of \bar{x} are also feasible for (CP). Since $c \neq 0$, there are points x in this neighbourhood with $c^T x < c^T \bar{x} \leq c^*$, which is impossible, since c^* is the optimal value of (CP).

Now let us make use of the following basic fact (see Section B.2.6):

Theorem 1.4.3 [Separation Theorem for Convex Sets] *Let S, T be nonempty non-intersecting convex subsets of a finite-dimensional Euclidean space \mathbf{E} with inner product $\langle \cdot, \cdot \rangle$. Then S and T can be separated by a linear functional: there exists a nonzero vector $\lambda \in \mathbf{E}$ such that*

$$\sup_{u \in S} \langle \lambda, u \rangle \leq \inf_{u \in T} \langle \lambda, u \rangle.$$

Applying the Separation Theorem to $S = M$ and $T = K$, we conclude that there exists $\lambda \in \mathbf{E}$ such that

$$\sup_{y \in M} \langle \lambda, y \rangle \leq \inf_{y \in \text{int } \mathbf{K}} \langle \lambda, y \rangle. \quad (1.4.4)$$

From the inequality it follows that the linear form $\langle \lambda, y \rangle$ of y is bounded below on $K = \text{int } \mathbf{K}$. Since this interior is a conic set:

$$y \in K, \mu > 0 \Rightarrow \mu y \in K$$

(why?), this boundedness implies that $\langle \lambda, y \rangle \geq 0$ for all $y \in K$. Consequently, $\langle \lambda, y \rangle \geq 0$ for all y from the closure of K , i.e., for all $y \in \mathbf{K}$. We conclude that $\lambda \geq_{\mathbf{K}_*} 0$, so that the inf in (1.4.4) is nonnegative. On the other hand, the infimum of a linear form over a conic set clearly cannot be positive; we conclude that the inf in (1.4.4) is 0, so that the inequality reads

$$\sup_{u \in M} \langle \lambda, u \rangle \leq 0.$$

Recalling the definition of M , we get

$$[A^* \lambda]^T x \leq \langle \lambda, b \rangle \quad (1.4.5)$$

for all x from the half-space $c^T x \leq c^*$. But the linear form $[A^* \lambda]^T x$ can be bounded above on the half-space if and only if the vector $A^* \lambda$ is proportional, with a nonnegative coefficient, to the vector c :

$$A^* \lambda = \mu c$$

for some $\mu \geq 0$. We claim that $\mu > 0$. Indeed, assuming $\mu = 0$, we get $A^* \lambda = 0$, whence $\langle \lambda, b \rangle \geq 0$ in view of (1.4.5). It is time now to recall that (CP) is strictly feasible, i.e., $A\bar{x} - b >_{\mathbf{K}} 0$ for some \bar{x} . Since $\lambda \geq_{\mathbf{K}_*} 0$ and $\lambda \neq 0$, the product $\langle \lambda, A\bar{x} - b \rangle$ should be strictly positive (why?), while in fact we know that the product is $-\langle \lambda, b \rangle \leq 0$ (since $A^* \lambda = 0$ and, as we have seen, $\langle \lambda, b \rangle \geq 0$).

Thus, $\mu > 0$. Setting $\lambda_* = \mu^{-1} \lambda$, we get

$$\begin{aligned} \lambda_* &\geq_{\mathbf{K}_*} 0 && [\text{since } \lambda \geq_{\mathbf{K}_*} 0 \text{ and } \mu > 0] \\ A^* \lambda_* &= c && [\text{since } A^* \lambda = \mu c] \\ c^T x &\leq \langle \lambda_*, b \rangle \quad \forall x : c^T x \leq c^* && [\text{see (1.4.5)}] \end{aligned}$$

We see that λ_* is feasible for (CD), the value of the dual objective at λ_* being at least c^* , as required.

It remains to consider the case $c = 0$. Here, of course, $c^* = 0$, and the existence of a dual feasible solution with the value of the objective $\geq c^* = 0$ is evident: the required solution is $\lambda = 0$. 3.a) is proved.

3.b): the result follows from 3.a) in view of the primal-dual symmetry.

4): Let x be primal feasible, and λ be dual feasible. Then

$$c^T x - \langle b, \lambda \rangle = (A^* \lambda)^T x - \langle b, \lambda \rangle = \langle Ax - b, \lambda \rangle.$$

We get a useful identity as follows:

(!) For every primal-dual feasible pair (x, λ) the duality gap $c^T x - \langle b, \lambda \rangle$ is equal to the inner product of the primal slack vector $y = Ax - b$ and the dual vector λ .

Note that (!) in fact does not require “full” primal-dual feasibility: x may be arbitrary (i.e., y should belong to the primal feasible plane $\text{Im } A - b$), and λ should belong to the dual feasible plane $A^* \lambda = c$, but y and λ not necessarily should belong to the respective cones.

In view of (!) the complementary slackness holds if and only if the duality gap is zero; thus, all we need is to prove 4.a).

The “primal residual” $c^T x - c^*$ and the “dual residual” $b^* - \langle b, \lambda \rangle$ are nonnegative, provided that x is primal feasible, and λ is dual feasible. It follows that the duality gap

$$c^T x - \langle b, \lambda \rangle = [c^T x - c^*] + [b^* - \langle b, \lambda \rangle] + [c^* - b^*]$$

is nonnegative (recall that $c^* \geq b^*$ by 2)), and it is zero if and only if $c^* = b^*$ and both primal and dual residuals are zero (i.e., x is primal optimal, and λ is dual optimal). All these arguments hold without any assumptions of strict feasibility. We see that the condition “the duality gap at a primal-dual feasible pair is zero” is always sufficient for primal-dual optimality of the pair; and if $c^* = b^*$, this sufficient condition is also necessary. Since in the case of 4) we indeed have $c^* = b^*$ (this is stated by 3)), 4.a) follows. ■

A useful consequence of the Conic Duality Theorem is the following

Corollary 1.4.2 Assume that both (CP) and (CD) are strictly feasible. Then both problems are solvable, the optimal values are equal to each other, and each one of the conditions 4.a), 4.b) is necessary and sufficient for optimality of a primal-dual feasible pair.

Indeed, by the Weak Conic Duality Theorem, if one of the problems is feasible, the other is bounded, and it remains to use the items 3) and 4) of the Conic Duality Theorem.

1.4.5.1 Refinement

We can slightly refine the Conic Duality Theorem, applying “special treatment” to scalar linear equality and inequality constraints. Specifically, assume that our primal problem is

$$\text{Opt}(P) = \min_x \left\{ c^T x : \begin{array}{l} A_1 x - b_1 \geq 0 \\ A_i x - b_i \in \mathbf{K}_i, 2 \leq i \leq m, \\ Rx = r \end{array} \right. \begin{array}{l} (a) \\ (b) \\ (c) \end{array} \right\} \quad (P)$$

where \mathbf{K}_i , $i = 2, 3, \dots, m$ are regular cones in Euclidean spaces. Applying the same “bounding mechanism” as when deriving the conic dual (CD) of (CP), we arrive at the dual problem

$$\text{Opt}(D) = \max_{z, \{y_i\}_{i=1}^m} \left\{ r^T z + b_1^T y_1 + \sum_{i=2}^m \langle b_i, y_i \rangle : \begin{array}{l} y_1 \geq 0, \\ y_i \in \mathbf{K}_i^*, 2 \leq i \leq m, \\ R^* z + \sum_{i=1}^m A_i^* y_i = c \end{array} \right\} \quad (D)$$

Essentially strict feasibility. Note that the structure of problem (D) is completely similar to the one of (P) – the variables, let them be called ξ , are subject to finitely many scalar linear equalities and inequalities and, on the top of it, finitely many conic inequalities $P_i \xi - p_i \in \mathbf{L}_i$, $i \in I$, where \mathbf{L}_i are regular cones in Euclidean spaces. Let us call such a conic problem *essentially strictly feasible*, if it admits a feasible solution $\bar{\xi}$ at which the conic inequalities are satisfied strictly: $P_i \bar{\xi} - p_i \in \text{int } \mathbf{L}_i$, $i \in I$. It turns out that the Conic Duality Theorem remains valid when one replaces in it “strict feasibility” with “essentially strict feasibility,” which is some progress: a strictly feasible conic problem clearly is essentially strictly feasible, but not necessarily vice versa. Thus, we intend to prove

Theorem 1.4.4 [Refined Conic Duality Theorem] *Consider a primal-dual pair of conic programs (P), (D). Then*

- (i) [Weak Duality] *One has $\text{Opt}(D) \leq \text{Opt}(P)$.*
- (ii) [Symmetry] *The duality is symmetric: (D) is a conic program, and the program dual to (D) is (equivalent to) (P).*
- (iii) [Refined Strong Duality] *If one of the programs (P), (D) is essentially strictly feasible and bounded, then the other program is solvable, and $\text{Opt}(P) = \text{Opt}(D)$.*

If both the programs are essentially strictly feasible, then both are solvable with equal optimal values.

Note that the Refined Conic Duality Theorem covers the usual Linear Programming Duality Theorem: the latter is the particular case $m = 1$ of the former.

For proof of Theorem 1.4.4, see. e.g., [23, Section 7.1.3].

1.4.6 Is something wrong with conic duality?

The statement of the Conic Duality Theorem is weaker than that of the LP Duality theorem: in the LP case, feasibility (even non-strict) and boundedness of either primal, or dual problem implies solvability of both the primal and the dual and equality between their optimal values. In the general conic case something “nontrivial” is stated only in the case of strict feasibility (and boundedness) of one of the problems. It can be demonstrated by examples that this phenomenon reflects the nature of things, and is not due to our ability to analyze it. The case of non-polyhedral cone \mathbf{K} is truly more complicated than the one of the nonnegative orthant \mathbf{K} ; as a result, a “word-by-word” extension of the LP Duality Theorem to the conic case is false.

Example 1.4.2 Consider the following conic problem with 2 variables $x = (x_1, x_2)^T$ and the 3-dimensional ice-cream cone \mathbf{K} :

$$\min \left\{ x_1 : Ax - b \equiv \begin{bmatrix} x_1 - x_2 \\ 1 \\ x_1 + x_2 \end{bmatrix} \geq_{\mathbf{L}^3} 0 \right\}.$$

Recalling the definition of \mathbf{L}^3 , we can write the problem equivalently as

$$\min \left\{ x_1 : \sqrt{(x_1 - x_2)^2 + 1} \leq x_1 + x_2 \right\},$$

i.e., as the problem

$$\min \{x_1 : 4x_1 x_2 \geq 1, x_1 + x_2 > 0\}.$$

Geometrically the problem is to minimize x_1 over the intersection of the 3D ice-cream cone with a 2D plane; the inverse image of this intersection in the “design plane” of variables x_1, x_2 is part of the 2D nonnegative orthant bounded by the hyperbola $x_1 x_2 \geq 1/4$. The problem is clearly strictly feasible (a strictly feasible solution is, e.g., $x = (1, 1)^T$) and bounded below, with the optimal value 0. This optimal value, however, is not achieved – the problem is unsolvable!

Example 1.4.3 Consider the following conic problem with two variables $x = (x_1, x_2)^T$ and the 3-dimensional ice-cream cone \mathbf{K} :

$$\min \left\{ x_2 : Ax - b = \begin{bmatrix} x_1 \\ x_2 \\ x_1 \end{bmatrix} \geq_{\mathbf{L}^3} 0 \right\}.$$

The problem is equivalent to the problem

$$\left\{ x_2 : \sqrt{x_1^2 + x_2^2} \leq x_1 \right\},$$

i.e., to the problem

$$\min \{x_2 : x_2 = 0, x_1 \geq 0\}.$$

The problem is clearly solvable, and its optimal set is the ray $\{x_1 \geq 0, x_2 = 0\}$.

Now let us build the conic dual to our (solvable!) primal. It is immediately seen that the cone dual to an ice-cream cone is this ice-cream cone itself. Thus, the dual problem is

$$\max_{\lambda} \left\{ 0 : \begin{bmatrix} \lambda_1 + \lambda_3 \\ \lambda_2 \end{bmatrix} = \begin{bmatrix} 0 \\ 1 \end{bmatrix}, \lambda \geq_{\mathbf{L}^3} 0 \right\}.$$

In spite of the fact that primal is solvable, the dual is infeasible: indeed, assuming that λ is dual feasible, we have $\lambda \geq_{\mathbf{L}^3} 0$, which means that $\lambda_3 \geq \sqrt{\lambda_1^2 + \lambda_2^2}$; since also $\lambda_1 + \lambda_3 = 0$, we come to $\lambda_2 = 0$, which contradicts the equality $\lambda_2 = 1$.

We see that the weakness of the Conic Duality Theorem as compared to the LP Duality one reflects pathologies which indeed may happen in the general conic case.

1.4.7 Consequences of the Conic Duality Theorem

1.4.7.1 Sufficient condition for infeasibility

Recall that a necessary and sufficient condition for infeasibility of a (finite) system of scalar linear inequalities (i.e., for a vector inequality with respect to the partial ordering \geq) is the possibility to combine these inequalities in a linear fashion in such a way that the resulting scalar linear inequality is contradictory. In the case of cone-generated vector inequalities a slightly weaker result can be obtained:

Proposition 1.4.2 [Conic Theorem on Alternative] *Consider a linear vector inequality*

$$Ax - b \geq_{\mathbf{K}} 0. \quad (\text{I})$$

(i) *If there exists λ satisfying*

$$\lambda \geq_{\mathbf{K}_*} 0, A^* \lambda = 0, \langle \lambda, b \rangle > 0, \quad (\text{II})$$

then (I) has no solutions.

(ii) *If (II) has no solutions, then (I) is “almost solvable” – for every positive ϵ there exists b' such that $\|b' - b\|_2 < \epsilon$ and the perturbed system*

$$Ax - b' \geq_{\mathbf{K}} 0$$

is solvable.

Moreover,

(iii) *(II) is solvable if and only if (I) is not “almost solvable”.*

Note the difference between the simple case when $\geq_{\mathbf{K}}$ is the usual partial ordering \geq and the general case. In the former, one can replace in (ii) “nearly solvable” by “solvable”; however, in the general conic case “almost” is unavoidable.

Example 1.4.4 Let system (I) be given by

$$Ax - b \equiv \begin{bmatrix} x+1 \\ x-1 \\ \sqrt{2}x \end{bmatrix} \geq_{\mathbf{L}^3} 0.$$

Recalling the definition of the ice-cream cone \mathbf{L}^3 , we can write the inequality equivalently as

$$\sqrt{2}x \geq \sqrt{(x+1)^2 + (x-1)^2} \equiv \sqrt{2x^2 + 2}, \quad (\text{i})$$

which of course is unsolvable. The corresponding system (II) is

$$\begin{aligned} \lambda_3 &\geq \sqrt{\lambda_1^2 + \lambda_2^2} && \left[\Leftrightarrow \lambda \geq_{\mathbf{L}_*^3} 0 \right] \\ \lambda_1 + \lambda_2 + \sqrt{2}\lambda_3 &= 0 && \left[\Leftrightarrow A^T \lambda = 0 \right] \\ \lambda_2 - \lambda_1 &> 0 && \left[\Leftrightarrow b^T \lambda > 0 \right] \end{aligned} \quad (\text{ii})$$

From the second of these relations, $\lambda_3 = -\frac{1}{\sqrt{2}}(\lambda_1 + \lambda_2)$, so that from the first inequality we get $0 \leq (\lambda_1 - \lambda_2)^2$, whence $\lambda_1 = \lambda_2$. But then the third inequality in (ii) is impossible! We see that here both (i) and (ii) have no solutions.

The geometry of the example is as follows. (i) asks to find a point in the intersection of the 3D ice-cream cone and a line. This line is an asymptote of the cone (it belongs to a 2D plane which crosses the cone in such way that the boundary of the cross-section is a branch of a hyperbola, and the line is one of two asymptotes of the hyperbola). Although the intersection is empty ((i) is unsolvable), small shifts of the line make the intersection nonempty (i.e., (i) is unsolvable and “almost solvable” at the same time). And it turns out that one cannot certify the fact that (i) itself is unsolvable by providing a solution to (ii).

Proof of the Proposition. (i) is evident (why?).

Let us prove (ii). To this end it suffices to verify that if (I) is not “almost solvable”, then (II) is solvable. Let us fix a vector $\sigma >_{\mathbf{K}} 0$ and look at the conic problem

$$\min_{x,t} \{t : Ax + t\sigma - b \geq_{\mathbf{K}} 0\} \quad (\text{CP})$$

in variables (x, t) . Clearly, the problem is strictly feasible (why?). Now, if (I) is not almost solvable, then, first, the matrix of the problem $[A; \sigma]$ satisfies the full column rank condition **A** (otherwise the image of the mapping $(x, t) \mapsto Ax + t\sigma - b$ would coincide with the image of the mapping $x \mapsto Ax - b$, which is not the case – the first of these images does intersect \mathbf{K} , while the second does not). Second, the optimal value in (CP) is strictly positive (otherwise the problem would admit feasible solutions with t close to 0, and this would mean that (I) is almost solvable). From the Conic Duality Theorem it follows that the dual problem of (CP)

$$\max_{\lambda} \{\langle b, \lambda \rangle : A^* \lambda = 0, \langle \sigma, \lambda \rangle = 1, \lambda \geq_{\mathbf{K}_*} 0\}$$

has a feasible solution with positive $\langle b, \lambda \rangle$, i.e., (II) is solvable.

It remains to prove (iii). Assume first that (I) is not almost solvable; then (II) must be solvable by (ii). Vice versa, assume that (II) is solvable, and let λ be a solution to (II). Then λ solves also all systems of the type (II) associated with small enough perturbations of b instead of b itself; by (i), it implies that all inequalities obtained from (I) by small enough perturbation of b are unsolvable. ■

1.4.7.2 When is a scalar linear inequality a consequence of a given linear vector inequality?

The question we are interested in is as follows: given a linear vector inequality

$$Ax \geq_{\mathbf{K}} b \quad (\text{V})$$

and a scalar inequality

$$c^T x \geq d \quad (\text{S})$$

we want to check whether (S) is a consequence of (V). If \mathbf{K} is the nonnegative orthant, the answer is given by the Inhomogeneous Farkas Lemma:

Inequality (S) is a consequence of a feasible system of linear inequalities $Ax \geq b$ if and only if (S) can be obtained from (V) and the trivial inequality $1 \geq 0$ in a linear fashion (by taking weighted sum with nonnegative weights).

In the general conic case we can get a slightly weaker result:

Proposition 1.4.3 (i) If (S) can be obtained from (V) and from the trivial inequality $1 \geq 0$ by admissible aggregation, i.e., there exist weight vector $\lambda \geq_{\mathbf{K}_*} 0$ such that

$$A^* \lambda = c, \langle \lambda, b \rangle \geq d,$$

then (S) is a consequence of (V).

(ii) If (S) is a consequence of a strictly feasible linear vector inequality (V), then (S) can be obtained from (V) by an admissible aggregation.

The difference between the case of the partial ordering \geq and a general partial ordering $\geq_{\mathbf{K}}$ is in the word “strictly” in (ii).

Proof of the proposition. (i) is evident (why?). To prove (ii), assume that (V) is strictly feasible and (S) is a consequence of (V) and consider the conic problem

$$\min_{x,t} \left\{ t : \bar{A} \begin{pmatrix} x \\ t \end{pmatrix} - \bar{b} \equiv \begin{bmatrix} Ax - b \\ d - c^T x + t \end{bmatrix} \geq_{\bar{\mathbf{K}}} 0 \right\},$$

$$\bar{\mathbf{K}} = \{(x, t) : x \in \mathbf{K}, t \geq 0\}$$

The problem is clearly strictly feasible (choose x to be a strictly feasible solution to (V) and then choose t to be large enough). The fact that (S) is a consequence of (V) says exactly that the optimal value in the problem is nonnegative. By the Conic Duality Theorem, the dual problem

$$\max_{\lambda,\mu} \left\{ \langle b, \lambda \rangle - d\mu : A^* \lambda - c = 0, \mu = 1, \begin{pmatrix} \lambda \\ \mu \end{pmatrix} \geq_{\bar{\mathbf{K}}_*} 0 \right\}$$

has a feasible solution with the value of the objective ≥ 0 . Since, as it is easily seen, $\bar{\mathbf{K}}_* = \{(\lambda, \mu) : \lambda \in \mathbf{K}_*, \mu \geq 0\}$, the indicated solution satisfies the requirements

$$\lambda \geq_{\mathbf{K}_*} 0, A^* \lambda = c, \langle b, \lambda \rangle \geq d,$$

i.e., (S) can be obtained from (V) by an admissible aggregation. ■

1.4.7.3 “Robust solvability status”

Examples 1.4.3 – 1.4.4 make it clear that in the general conic case we may meet “pathologies” which do not occur in LP. E.g., a feasible and bounded problem may be unsolvable, the dual to a solvable conic problem may be infeasible, etc. Where the pathologies come from? Looking at our “pathological examples”, we arrive at the following guess: the source of the pathologies is that *in these examples, the “solvability status” of the primal problem is non-robust – it can be changed by small perturbations of the data*. This issue of robustness is very important in modelling, and it deserves a careful investigation.

Data of a conic problem. When asked “What are the data of an LP program $\min\{c^T x : Ax - b \geq 0\}$ ”, everybody will give the same answer: “the objective c , the constraint matrix A and the right hand side vector b ”. Similarly, for a conic problem

$$\min \left\{ c^T x : Ax - b \geq_{\mathbf{K}} 0 \right\}, \quad (\text{CP})$$

its data, *by definition*, is the triple (c, A, b) , while the sizes of the problem – the dimension n of x and the dimension m of \mathbf{K} , same as the underlying cone \mathbf{K} itself, are considered as the structure of (CP).

Robustness. A question of primary importance is whether the properties of the program (CP) (feasibility, solvability, etc.) are stable with respect to perturbations of the data. The reasons which make this question important are as follows:

- In actual applications, especially those arising in Engineering, the data are normally inexact: their true values, even when they “exist in the nature”, are not known exactly when the problem is processed. Consequently, the results of the processing say something definite about the “true” problem only if these results are robust with respect to small data perturbations i.e., the properties of (CP) we have discovered are shared not only by the particular (“nominal”) problem we were processing, but also by all problems with nearby data.

- Even when the exact data are available, we should take into account that processing them computationally we unavoidably add “noise” like rounding errors (you simply cannot load something like $1/7$ to the standard computer). As a result, a real-life computational routine can recognize only those properties of the input problem which are stable with respect to small perturbations of the data.

Due to the above reasons, we should study not only whether a given problem (CP) is feasible/bounded/solvable, etc., but also whether these properties are robust – remain unchanged under small data perturbations. As it turns out, the Conic Duality Theorem allows to recognize “robust feasibility/boundedness/solvability...”.

Let us start with introducing the relevant concepts. We say that (CP) is

- *robust feasible*, if all “sufficiently close” problems (i.e., those of the same structure (n, m, \mathbf{K}) and with data close enough to those of (CP)) are feasible;
- *robust infeasible*, if all sufficiently close problems are infeasible;
- *robust bounded below*, if all sufficiently close problems are bounded below (i.e., their objectives are bounded below on their feasible sets);
- *robust unbounded*, if all sufficiently close problems are not bounded;
- *robust solvable*, if all sufficiently close problems are solvable.

Note that a problem which is not robust feasible, *not necessarily* is robust infeasible, since among close problems there may be both feasible and infeasible (look at Example 1.4.3 – slightly shifting and rotating the plane $\text{Im } A - b$, we may get whatever we want – a feasible bounded problem, a feasible unbounded problem, an infeasible problem...). This is why we need two kinds of definitions: one of “robust presence of a property” and one more of “robust absence of the same property”.

Now let us look what are necessary and sufficient conditions for the most important robust forms of the “solvability status”.

Proposition 1.4.4 [Robust feasibility] (CP) is robust feasible if and only if it is strictly feasible, in which case the dual problem (D) is robust bounded above.

Proof. The statement is nearly tautological. Let us fix $\delta >_{\mathbf{K}} 0$. If (CP) is robust feasible, then for small enough $t > 0$ the perturbed problem $\min\{c^T x : Ax - b - t\delta \geq_{\mathbf{K}} 0\}$ should be feasible; a feasible solution to the perturbed problem clearly is a strictly feasible solution to (CP). The inverse implication is evident (a strictly feasible solution to (CP) remains feasible for all problems with close enough data). It remains to note that if all problems sufficiently close to (CP) are feasible, then their duals, by the Weak Conic Duality Theorem, are bounded above, so that (D) is robust above bounded. ■

Proposition 1.4.5 [Robust infeasibility] (CP) is robust infeasible if and only if the system

$$\langle b, \lambda \rangle = 1, A^* \lambda = 0, \lambda \geq_{\mathbf{K}_*} 0$$

is robust feasible, or, which is the same (by Proposition 1.4.4), if and only if the system

$$\langle b, \lambda \rangle = 1, A^* \lambda = 0, \lambda >_{\mathbf{K}_*} 0 \tag{1.4.6}$$

has a solution.

Proof. First assume that (1.4.6) is solvable, and let us prove that all problems sufficiently close to (CP) are infeasible. Let us fix a solution $\bar{\lambda}$ to (1.4.6). Since A is of full column rank, simple Linear Algebra says that the systems $[A']^* \lambda = 0$ are solvable for all matrices A' from a small enough neighbourhood U of A ; moreover, the corresponding solution $\lambda(A')$ can be chosen to satisfy $\lambda(A) = \bar{\lambda}$ and to be continuous in $A' \in U$. Since $\lambda(A')$ is continuous and $\lambda(A) >_{\mathbf{K}_*} 0$, we have $\lambda(A')$ is $>_{\mathbf{K}_*} 0$ in a neighbourhood of A ; shrinking U appropriately, we may assume that $\lambda(A') >_{\mathbf{K}_*} 0$ for all $A' \in U$. Now, $b^T \bar{\lambda} = 1$; by continuity reasons, there exists a neighbourhood V of b and a neighbourhood U' of A such that $b' \in V$ and all $A' \in U'$ one has $\langle b', \lambda(A') \rangle > 0$.

Thus, we have seen that there exist a neighbourhood U' of A and a neighbourhood V of b , along with a function $\lambda(A')$, $A' \in U'$, such that

$$\langle b', \lambda(A') \rangle > 0, [A']^* \lambda(A') = 0, \lambda(A') \geq_{\mathbf{K}_*} 0$$

for all $b' \in V$ and $A' \in U'$. By Proposition 1.4.2.(i) it means that all the problems

$$\min \{[c']^T x : A'x - b' \geq_{\mathbf{K}} 0\}$$

with $b' \in V$ and $A' \in U'$ are infeasible, so that (CP) is robust infeasible.

Now let us assume that (CP) is robust infeasible, and let us prove that then (1.4.6) is solvable. Indeed, by the definition of robust infeasibility, there exist neighbourhoods U of A and V of b such that all vector inequalities

$$A'x - b' \geq_{\mathbf{K}} 0$$

with $A' \in U$ and $b' \in V$ are unsolvable. It follows that whenever $A' \in U$ and $b' \in V$, the vector inequality

$$A'x - b' \geq_{\mathbf{K}} 0$$

is not almost solvable (see Proposition 1.4.2). We conclude from Proposition 1.4.2.(ii) that for every $A' \in U$ and $b' \in V$ there exists $\lambda = \lambda(A', b')$ such that

$$\langle b', \lambda(A', b') \rangle > 0, [A']^* \lambda(A', b') = 0, \lambda(A', b') \geq_{\mathbf{K}_*} 0.$$

Now let us choose $\lambda_0 >_{\mathbf{K}_*} 0$. For all small enough positive ϵ we have $A_\epsilon = A + \epsilon b[A^* \lambda_0]^T \in U$. Let us choose an ϵ with the latter property to be so small that $\epsilon \langle b, \lambda_0 \rangle > -1$ and set $A' = A_\epsilon$, $b' = b$. According to the previous observation, there exists $\lambda = \lambda(A', b)$ such that

$$\langle b, \lambda \rangle > 0, [A']^* \lambda \equiv A^*[\lambda + \epsilon \langle b, \lambda \rangle \lambda_0] = 0, \lambda \geq_{\mathbf{K}_*} 0.$$

Setting $\bar{\lambda} = \lambda + \epsilon \langle b, \lambda \rangle \lambda_0$, we get $\bar{\lambda} >_{\mathbf{K}_*} 0$ (since $\lambda \geq_{\mathbf{K}_*} 0$, $\lambda_0 >_{\mathbf{K}_*} 0$ and $\langle b, \lambda \rangle > 0$), while $A^* \bar{\lambda} = 0$ and $\langle b, \bar{\lambda} \rangle = \langle b, \lambda \rangle(1 + \epsilon \langle b, \lambda_0 \rangle) > 0$. Multiplying $\bar{\lambda}$ by appropriate positive factor, we get a solution to (1.4.6).

■

Now we are able to formulate our main result on “robust solvability”.

Proposition 1.4.6 *For a conic problem (CP) the following conditions are equivalent to each other*

- (i) (CP) is robust feasible and robust bounded (below);
- (ii) (CP) is robust solvable;
- (iii) (D) is robust solvable;
- (iv) (D) is robust feasible and robust bounded (above);
- (v) Both (CP) and (D) are strictly feasible.

In particular, under every one of these equivalent assumptions, both (CP) and (D) are solvable with equal optimal values.

Proof. (i) \Rightarrow (v): If (CP) is robust feasible, it also is strictly feasible (Proposition 1.4.4). If, in addition, (CP) is robust bounded below, then (D) is robust solvable (by the Conic Duality Theorem); in particular, (D) is robust feasible and therefore strictly feasible (again Proposition 1.4.4).

(v) \Rightarrow (ii): The implication is given by the Conic Duality Theorem.

(ii) \Rightarrow (i): trivial.

We have proved that (i) \equiv (ii) \equiv (v). Due to the primal-dual symmetry, we also have proved that (iii) \equiv (iv) \equiv (v). ■

1.5 Exercises

1.5.1 Around General Theorem on Alternative

Exercise 1.4 Derive General Theorem on Alternative from Homogeneous Farkas Lemma

Hint: Verify that the system

$$(\mathcal{S}) : \quad \begin{cases} a_i^T x > b_i, & i = 1, \dots, m_s, \\ a_i^T x \geq b_i, & i = m_s + 1, \dots, m. \end{cases}$$

in variables x has no solution if and only if the homogeneous inequality

$$\epsilon \leq 0$$

in variables x, ϵ, t is a consequence of the system of homogeneous inequalities

$$\begin{cases} a_i^T x - b_i t - \epsilon \geq 0, & i = 1, \dots, m_s, \\ a_i^T x - b_i t \geq 0, & i = m_s + 1, \dots, m, \\ t \geq \epsilon, \end{cases}$$

in these variables.

There exist several particular cases of GTA (which in fact are equivalent to GTA); the goal of the next exercise is to prove the corresponding statements.

Exercise 1.5 Derive the following statements from the General Theorem on Alternative:

1. [Gordan's Theorem on Alternative] One of the inequality systems

$$(I) \quad Ax < 0, \quad x \in \mathbf{R}^n,$$

$$(II) \quad A^T y = 0, \quad 0 \neq y \geq 0, \quad y \in \mathbf{R}^m,$$

(A being an $m \times n$ matrix, x are variables in (I), y are variables in (II)) has a solution if and only if the other one has no solutions.

2. [Inhomogeneous Farkas Lemma] A linear inequality in variables x

$$a^T x \leq p \tag{1.5.1}$$

is a consequence of a solvable system of linear inequalities

$$Nx \leq q \tag{1.5.2}$$

if and only if it is a "linear consequence" of the system and the trivial inequality

$$0^T x \leq 1,$$

i.e., if it can be obtained by taking weighted sum, with nonnegative coefficients, of the inequalities from the system and this trivial inequality.

Algebraically: (1.5.1) is a consequence of solvable system (1.5.2) if and only if

$$a = N^T \nu$$

for some nonnegative vector ν such that

$$\nu^T q \leq p.$$

3. [Motzkin's Theorem on Alternative] The system

$$Sx < 0, \quad Nx \leq 0$$

in variables x has no solutions if and only if the system

$$S^T \sigma + N^T \nu = 0, \quad \sigma \geq 0, \quad \nu \geq 0, \quad \sigma \neq 0$$

in variables σ, ν has a solution.

Exercise 1.6 Consider the linear inequality

$$x + y \leq 2$$

and the system of linear inequalities

$$\begin{cases} x \leq 1 \\ -x \leq -100 \end{cases}$$

Our inequality clearly is a consequence of the system – it is satisfied at every solution to it (simply because there are no solutions to the system at all). According to the Inhomogeneous Farkas Lemma, the inequality should be a linear consequence of the system and the trivial inequality $0 \leq 1$, i.e., there should exist nonnegative ν_1, ν_2 such that

$$\begin{pmatrix} 1 \\ 1 \end{pmatrix} = \nu_1 \begin{pmatrix} 1 \\ 0 \end{pmatrix} + \nu_2 \begin{pmatrix} -1 \\ 0 \end{pmatrix}, \quad \nu_1 - 100\nu_2 \leq 2,$$

which clearly is not the case. What is the reason for the observed “contradiction”?

1.5.2 Around cones

Attention! In what follows, if otherwise is not explicitly stated, “cone” is a shorthand for regular (i.e., closed pointed cone with a nonempty interior) cone, \mathbf{K} denotes a cone, and \mathbf{K}_* is the cone dual to \mathbf{K} .

Exercise 1.7 Let \mathbf{K} be a cone, and let $\bar{x} >_{\mathbf{K}} 0$. Prove that $x >_{\mathbf{K}} 0$ if and only if there exists positive real t such that $x \geq_{\mathbf{K}} t\bar{x}$.

Exercise 1.8 1) Prove that if $0 \neq x \geq_{\mathbf{K}} 0$ and $\lambda >_{\mathbf{K}_*} 0$, then $\lambda^T x > 0$.

2) Assume that $\lambda \geq_{\mathbf{K}_*} 0$. Prove that $\lambda >_{\mathbf{K}_*} 0$ if and only if $\lambda^T x > 0$ whenever $0 \neq x \geq_{\mathbf{K}} 0$.

3) Prove that $\lambda >_{\mathbf{K}_*} 0$ if and only if the set

$$\{x \geq_{\mathbf{K}} 0 : \lambda^T x \leq 1\}$$

is compact.

1.5.2.1 Calculus of cones

Exercise 1.9 Prove the following statements:

1) [stability with respect to direct multiplication] Let $\mathbf{K}_i \subset \mathbf{R}^{n_i}$ be cones, $i = 1, \dots, k$. Prove that the direct product of the cones:

$$\mathbf{K} = \mathbf{K}_1 \times \dots \times \mathbf{K}_k = \{(x_1, \dots, x_k) : x_i \in \mathbf{K}_i, i = 1, \dots, k\}$$

is a cone in $\mathbf{R}^{n_1+\dots+n_k} = \mathbf{R}^{n_1} \times \dots \times \mathbf{R}^{n_k}$.

Prove that the cone dual to \mathbf{K} is the direct product of the cones dual to \mathbf{K}_i , $i = 1, \dots, k$.

2) [stability with respect to taking inverse image] Let \mathbf{K} be a cone in \mathbf{R}^n and $u \mapsto Au$ be a linear mapping from certain \mathbf{R}^k to \mathbf{R}^n with trivial null space ($\text{Null}(A) = \{0\}$) and such that $\text{Im } A \cap \text{int } \mathbf{K} \neq \emptyset$. Prove that the inverse image of \mathbf{K} under the mapping:

$$A^{-1}(\mathbf{K}) = \{u : Au \in \mathbf{K}\}$$

is a cone in \mathbf{R}^k .

Prove that the cone dual to $A^{-1}(\mathbf{K})$ is $A^T \mathbf{K}_*$, i.e.

$$(A^{-1}(\mathbf{K}))_* = \{A^T \lambda : \lambda \in \mathbf{K}_*\}.$$

3) [stability with respect to taking linear image] Let \mathbf{K} be a cone in \mathbf{R}^n and $y = Ax$ be a linear mapping from \mathbf{R}^n onto \mathbf{R}^N (i.e., the image of A is the entire \mathbf{R}^N). Assume $\text{Null}(A) \cap \mathbf{K} = \{0\}$.

Prove that then the set

$$A\mathbf{K} = \{Ax : x \in \mathbf{K}\}$$

is a cone in \mathbf{R}^N .

Prove that the cone dual to $A\mathbf{K}$ is

$$(A\mathbf{K})_* = \{\lambda \in \mathbf{R}^N : A^T \lambda \in \mathbf{K}_*\}.$$

Demonstrate by example that if in the above statement the assumption $\text{Null}(A) \cap \mathbf{K} = \{0\}$ is weakened to $\text{Null}(A) \cap \text{int } \mathbf{K} = \emptyset$, then the set $A(\mathbf{K})$ may happen to be non-closed.

Hint. Look what happens when the 3D ice-cream cone is projected onto its tangent plane.

1.5.2.2 Primal-dual pairs of cones and orthogonal pairs of subspaces

Exercise 1.10 Let A be a $m \times n$ matrix of full column rank and \mathbf{K} be a cone in \mathbf{R}^m .

1) Prove that at least one of the following facts always takes place:

- (i) There exists a nonzero $x \in \text{Im } A$ which is $\geq_{\mathbf{K}} 0$;
- (ii) There exists a nonzero $\lambda \in \text{Null}(A^T)$ which is $\geq_{\mathbf{K}_*} 0$.

Geometrically: given a primal-dual pair of cones \mathbf{K}, \mathbf{K}_* and a pair L, L^\perp of linear subspaces which are orthogonal complements of each other, we either can find a nontrivial ray in the intersection $L \cap \mathbf{K}$, or in the intersection $L^\perp \cap \mathbf{K}_*$, or both.

2) Prove that there exists $\lambda \in \text{Null}(A^T)$ which is $>_{\mathbf{K}_*} 0$ (this is the strict version of (ii)) if and only if (i) is false. Prove that, similarly, there exists $x \in \text{Im } A$ which is $>_{\mathbf{K}} 0$ (this is the strict version of (i)) if and only if (ii) is false.

Geometrically: if \mathbf{K}, \mathbf{K}_* is a primal-dual pair of cones and L, L^\perp are linear subspaces which are orthogonal complements of each other, then the intersection $L \cap \mathbf{K}$ is trivial (i.e., is the singleton $\{0\}$) if and only if the intersection $L^\perp \cap \text{int } \mathbf{K}_*$ is nonempty.

1.5.2.3 Several interesting cones

Given a cone \mathbf{K} along with its dual \mathbf{K}_* , let us call a *complementary pair* every pair $x \in \mathbf{K}$, $\lambda \in \mathbf{K}_*$ such that

$$\lambda^T x = 0.$$

Recall that in “good cases” (e.g., under the premise of item 4 of the Conic Duality Theorem) a pair of feasible solutions (x, λ) of a primal-dual pair of conic problems

$$\min \left\{ c^T x : Ax - b \geq_{\mathbf{K}} 0 \right\}$$

$$\max \left\{ b^T \lambda : A^T \lambda = c, \lambda \geq_{\mathbf{K}_*} 0 \right\}$$

is primal-dual optimal if and only if the “primal slack” $y = Ax - b$ and λ are complementary.

Exercise 1.11 [Nonnegative orthant] *Prove that the n -dimensional nonnegative orthant \mathbf{R}_+^n is a cone and that it is self-dual:*

$$(\mathbf{R}_+^n)_* = \mathbf{R}_+^n.$$

What are complementary pairs?

Exercise 1.12 [Ice-cream cone] *Let \mathbf{L}^n be the n -dimensional ice-cream cone:*

$$\mathbf{L}^n = \{x \in \mathbf{R}^n : x_n \geq \sqrt{x_1^2 + \dots + x_{n-1}^2}\}.$$

- 1) *Prove that \mathbf{L}^n is a cone.*
- 2) *Prove that the ice-cream cone is self-dual:*

$$(\mathbf{L}^n)_* = \mathbf{L}^n.$$

- 3) *Characterize the complementary pairs.*

Exercise 1.13 [Positive semidefinite cone] *Let \mathbf{S}_+^n be the cone of $n \times n$ positive semidefinite matrices in the space \mathbf{S}^n of symmetric $n \times n$ matrices. Assume that \mathbf{S}^n is equipped with the Frobenius inner product*

$$\langle X, Y \rangle = \text{Tr}(XY) = \sum_{i,j=1}^n X_{ij}Y_{ij}.$$

- 1) *Prove that \mathbf{S}_+^n indeed is a cone.*
- 2) *Prove that the semidefinite cone is self-dual:*

$$(\mathbf{S}_+^n)_* = \mathbf{S}_+^n,$$

i.e., that the Frobenius inner products of a symmetric matrix Λ with all positive semidefinite matrices X of the same size are nonnegative if and only if the matrix Λ itself is positive semidefinite.

- 3) *Prove the following characterization of the complementary pairs:*

Two matrices $X \in \mathbf{S}_+^n$, $\Lambda \in (\mathbf{S}_+^n)_ \equiv \mathbf{S}_+^n$ are complementary (i.e., $\langle \Lambda, X \rangle = 0$) if and only if their matrix product is zero: $\Lambda X = X \Lambda = 0$. In particular, matrices from a complementary pair commute and therefore share a common orthonormal eigenbasis.*

1.5.3 Around conic problems

1.5.3.1 Several primal-dual pairs

Exercise 1.14 [The min-max Steiner problem] Consider the problem as follows:

Given N points b_1, \dots, b_N in \mathbf{R}^n , find a point $x \in \mathbf{R}^n$ which minimizes the maximum (Euclidean) distance from itself to the points b_1, \dots, b_N , i.e., solve the problem

$$\min_x \max_{i=1, \dots, N} \|x - b_i\|_2.$$

Imagine, e.g., that $n = 2$, b_1, \dots, b_N are locations of villages and you are interested to locate a fire station for which the worst-case distance to a possible fire is as small as possible.

- 1) Pose the problem as a conic quadratic one – a conic problem associated with a direct product of ice-cream cones.
- 2) Build the dual problem.
- 3) What is the geometric interpretation of the dual? Are the primal and the dual strictly feasible? Solvable? With equal optimal values? What is the meaning of the complementary slackness?

Exercise 1.15 [The weighted Steiner problem] Consider the problem as follows:

Given N points b_1, \dots, b_N in \mathbf{R}^n along with positive weights ω_i , $i = 1, \dots, N$, find a point $x \in \mathbf{R}^n$ which minimizes the weighted sum of its (Euclidean) distances to the points b_1, \dots, b_N , i.e., solve the problem

$$\min_x \sum_{i=1}^N \omega_i \|x - b_i\|_2.$$

Imagine, e.g., that $n = 2$, b_1, \dots, b_N are locations of N villages and you are interested to place a telephone station for which the total cost of cables linking the station and the villages is as small as possible. The weights can be interpreted as the per mile cost of the cables (they may vary from village to village due to differences in populations and, consequently, in the required capacities of the cables).

- 1) Pose the problem as a conic quadratic one.
- 2) Build the dual problem.
- 3) What is the geometric interpretation of the dual? Are the primal and the dual strictly feasible? Solvable? With equal optimal values? What is the meaning of the complementary slackness?

1.5.4 Feasible and level sets of conic problems

Attention! Remember that by our Assumption A matrix A below is of full column rank!

Consider a feasible conic problem

$$\min \left\{ c^T x : Ax - b \geq_{\mathbf{K}} 0 \right\}. \quad (\text{CP})$$

In many cases it is important to know whether the problem has

- 1) bounded feasible set $\{x : Ax - b \geq_{\mathbf{K}} 0\}$
- 2) bounded level sets

$$\{x : Ax - b \geq_{\mathbf{K}} 0, c^T x \leq a\}$$

for all real a .

Exercise 1.16 Let (CP) be feasible. Then the following four properties are equivalent:

- (i) the feasible set of the problem is bounded;
- (ii) the set of primal slacks $Y = \{y : y \geq_{\mathbf{K}} 0, y = Ax - b\}$ is bounded.
- (iii) $\text{Im } A \cap \mathbf{K} = \{0\}$;
- (iv) the system of vector (in)equalities

$$A^T \lambda = 0, \lambda >_{\mathbf{K}_*} 0$$

is solvable.

Corollary. The property of (CP) to have a bounded feasible set is independent of the particular value of b , provided that with this b (CP) is feasible!

Exercise 1.17 Let problem (CP) be feasible. Prove that the following two conditions are equivalent:

- (i) (CP) has bounded level sets;
- (ii) The dual problem

$$\max \left\{ b^T \lambda : A^T \lambda = c, \lambda \geq_{\mathbf{K}_*} 0 \right\}$$

is strictly feasible.

Corollary. The property of (CP) to have bounded level sets is independent of the particular value of b , provided that with this b (CP) is feasible!

Lecture 2

Conic Quadratic Programming

Several “generic” families of conic problems are of special interest, both from the viewpoint of theory and applications. The cones underlying these problems are simple enough, so that one can describe explicitly the dual cone; as a result, the general duality machinery we have developed becomes “algorithmic”, as in the Linear Programming case. Moreover, in many cases this “algorithmic duality machinery” allows to understand more deeply the original model, to convert it into equivalent forms better suited for numerical processing, etc. The relative simplicity of the underlying cones also enables one to develop efficient computational methods for the corresponding conic problems. The most famous example of a “nice” generic conic problem is, doubtless, Linear Programming; however, it is not the only problem of this sort. Two other nice generic conic problems of extreme importance are *Conic Quadratic* and *Semidefinite* programs. We are about to consider the first of these two problems.

2.1 Conic Quadratic problems: preliminaries

Recall the definition of the m -dimensional ice-cream (\equiv second-order \equiv Lorentz) cone \mathbf{L}^m :

$$\mathbf{L}^m = \{x = (x_1, \dots, x_m) \in \mathbf{R}^m : x_m \geq \sqrt{x_1^2 + \dots + x_{m-1}^2}\}, \quad m \geq 2.$$

A *conic quadratic problem* is a conic problem

$$\min_x \left\{ c^T x : Ax - b \geq_{\mathbf{K}} 0 \right\} \tag{CP}$$

for which the cone \mathbf{K} is a direct product of several ice-cream cones:

$$\begin{aligned} \mathbf{K} &= \mathbf{L}^{m_1} \times \mathbf{L}^{m_2} \times \dots \times \mathbf{L}^{m_k} \\ &= \left\{ y = \begin{pmatrix} y[1] \\ y[2] \\ \vdots \\ y[k] \end{pmatrix} : y[i] \in \mathbf{L}^{m_i}, i = 1, \dots, k \right\}. \end{aligned} \tag{2.1.1}$$

In other words, a conic quadratic problem is an optimization problem with linear objective and finitely many “ice-cream constraints”

$$A_i x - b_i \geq_{\mathbf{L}^{m_i}} 0, \quad i = 1, \dots, k,$$

where

$$[A; b] = \begin{bmatrix} [A_1; b_1] \\ [A_2; b_2] \\ \vdots \\ [A_k; b_k] \end{bmatrix}$$

is the partition of the data matrix $[A; b]$ corresponding to the partition of y in (2.1.1). Thus, a conic quadratic program can be written as

$$\min_x \left\{ c^T x : A_i x - b_i \geq_{\mathbf{L}^{m_i}} 0, i = 1, \dots, k \right\}. \quad (2.1.2)$$

Recalling the definition of the relation $\geq_{\mathbf{L}^m}$ and partitioning the data matrix $[A_i, b_i]$ as

$$[A_i; b_i] = \begin{bmatrix} D_i & d_i \\ p_i^T & q_i \end{bmatrix}$$

where D_i is of the size $(m_i - 1) \times \dim x$, we can write down the problem as

$$\min_x \left\{ c^T x : \|D_i x - d_i\|_2 \leq p_i^T x - q_i, i = 1, \dots, k \right\}; \quad (\text{QP})$$

this is the ‘‘most explicit’’ form is the one we prefer to use. In this form, D_i are matrices of the same row dimension as x , d_i are vectors of the same dimensions as the column dimensions of the matrices D_i , p_i are vectors of the same dimension as x and q_i are reals.

It is immediately seen that (2.1.1) is indeed a cone, in fact a self-dual one: $\mathbf{K}_* = \mathbf{K}$. Consequently, the problem dual to (CP) is

$$\max_{\lambda} \left\{ b^T \lambda : A^T \lambda = c, \lambda \geq_{\mathbf{K}} 0 \right\}.$$

Denoting $\lambda = \begin{pmatrix} \lambda_1 \\ \lambda_2 \\ \vdots \\ \lambda_k \end{pmatrix}$ with m_i -dimensional blocks λ_i (cf. (2.1.1)), we can write the dual problem as

$$\max_{\lambda_1, \dots, \lambda_m} \left\{ \sum_{i=1}^k b_i^T \lambda_i : \sum_{i=1}^k A_i^T \lambda_i = c, \lambda_i \geq_{\mathbf{L}^{m_i}} 0, i = 1, \dots, k \right\}.$$

Recalling the meaning of $\geq_{\mathbf{L}^{m_i}} 0$ and representing $\lambda_i = \begin{pmatrix} \mu_i \\ \nu_i \end{pmatrix}$ with scalar component ν_i , we finally come to the following form of the problem dual to (QP):

$$\max_{\mu_i, \nu_i} \left\{ \sum_{i=1}^k [\mu_i^T d_i + \nu_i q_i] : \sum_{i=1}^k [D_i^T \mu_i + \nu_i p_i] = c, \|\mu_i\|_2 \leq \nu_i, i = 1, \dots, k \right\}. \quad (\text{QD})$$

The design variables in (QD) are vectors μ_i of the same dimensions as the vectors d_i and reals $\nu_i, i = 1, \dots, k$.

Since from now on we will treat (QP) and (QD) as the standard forms of a conic quadratic problem and its dual, we now interpret for these two problems our basic assumption **A** from Lecture 2 and notions like feasibility, strict feasibility, boundedness, etc. Assumption **A** now reads (why?):

There is no nonzero x which is orthogonal to all rows of all matrices D_i and to all vectors p_i , $i = 1, \dots, k$

and we always make this assumption by default. Now, among notions like feasibility, solvability, etc., the only notion which does need an interpretation is strict feasibility, which now reads as follows (why?):

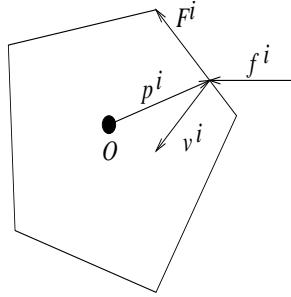
Strict feasibility of (QP) means that there exist \bar{x} such that $\|D_i \bar{x} - d_i\|_2 < p_i^T \bar{x} - q_i$ for all i .

Strict feasibility of (QD) means that there exists a feasible solution $\{\bar{\mu}_i, \bar{\nu}_i\}_{i=1}^k$ to the problem such that $\|\bar{\mu}_i\|_2 < \bar{\nu}_i$ for all $i = 1, \dots, k$.

2.2 Examples of conic quadratic problems

2.2.1 Contact problems with static friction [19]

Consider a rigid body in \mathbf{R}^3 and a robot with N fingers. When can the robot hold the body? To pose the question mathematically, let us look what happens at the point p^i of the body which is in contact with i -th finger of the robot:



Geometry of i -th contact

[p^i is the contact point; f^i is the contact force; v^i is the inward normal to the surface]

Let v^i be the unit inward normal to the surface of the body at the point p^i where i -th finger touches the body, f^i be the contact force exerted by i -th finger, and F^i be the friction force caused by the contact. Physics (Coulomb's law) says that the latter force is tangential to the surface of the body:

$$(F^i)^T v^i = 0 \quad (2.2.1)$$

and its magnitude cannot exceed μ times the magnitude of the normal component of the contact force, where μ is the friction coefficient:

$$\|F^i\|_2 \leq \mu (f^i)^T v^i. \quad (2.2.2)$$

Assume that the body is subject to additional external forces (e.g., gravity); as far as their mechanical consequences are concerned, all these forces can be represented by a single force – their sum – F^{ext} along with the torque T^{ext} – the sum of vector products of the external forces and the points where they are applied.

In order for the body to be in static equilibrium, the total force acting at the body and the total torque should be zero:

$$\begin{aligned} \sum_{i=1}^N (f^i + F^i) + F^{\text{ext}} &= 0 \\ \sum_{i=1}^N p^i \times (f^i + F^i) + T^{\text{ext}} &= 0, \end{aligned} \quad (2.2.3)$$

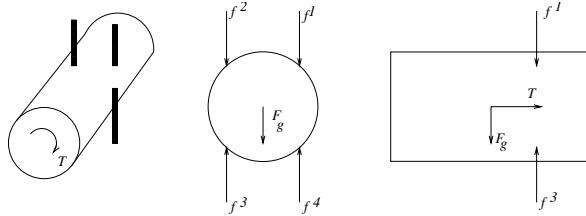
where $p \times q$ stands for the vector product of two 3D vectors p and q ¹⁾.

The question “whether the robot is capable to hold the body” can be interpreted as follows. Assume that $f^i, F^{\text{ext}}, T^{\text{ext}}$ are given. If the friction forces F^i can adjust themselves to satisfy the friction constraints (2.2.1) – (2.2.2) and the equilibrium equations (2.2.3), i.e., if the system of constraints (2.2.1), (2.2.2), (2.2.3) with respect to unknowns F^i is solvable, then, and only then, the robot holds the body (“the body is in a *stable grasp*”).

Thus, the question of stable grasp is the question of solvability of the system (S) of constraints (2.2.1), (2.2.2) and (2.2.3), which is a system of conic quadratic and linear constraints in the variables $f^i, F^{\text{ext}}, T^{\text{ext}}, \{F^i\}$. It follows that typical grasp-related optimization problems can be posed as CQPs. Here is an example:

The robot should hold a cylinder by four fingers, all acting in the vertical direction.

The external forces and torques acting at the cylinder are the gravity F_g and an externally applied torque T along the cylinder axis, as shown in the picture:



Perspective, front and side views

The magnitudes ν_i of the forces f_i may vary in a given segment $[0, F_{\max}]$.

What can be the largest magnitude τ of the external torque T such that a stable grasp is still possible?

Denoting by u^i the directions of the fingers, by v^i the directions of the inward normals to cylinder’s surface at the contact points, and by u the direction of the axis of the cylinder, we

¹⁾Here is the definition: if $p = \begin{pmatrix} p_1 \\ p_2 \\ p_3 \end{pmatrix}$, $q = \begin{pmatrix} q_1 \\ q_2 \\ q_3 \end{pmatrix}$ are two 3D vectors, then

$$[p, q] = \begin{pmatrix} \text{Det} \begin{pmatrix} p_2 & p_3 \\ q_2 & q_3 \end{pmatrix} \\ \text{Det} \begin{pmatrix} p_3 & p_1 \\ q_3 & q_1 \end{pmatrix} \\ \text{Det} \begin{pmatrix} p_1 & p_2 \\ q_1 & q_2 \end{pmatrix} \end{pmatrix}$$

The vector $[p, q]$ is orthogonal to both p and q , and $\|[p, q]\|_2 = \|p\|_2 \|q\|_2 \sin(\hat{pq})$.

can pose the problem as the optimization program

$$\begin{aligned}
 & \max \tau \\
 \text{s.t.} \\
 & \sum_{i=1}^4 (\nu_i u^i + F^i) + F_g = 0 && [\text{total force equals 0}] \\
 & \sum_{i=1}^4 p^i \times (\nu_i u^i + F^i) + \tau u = 0 && [\text{total torque equals 0}] \\
 & (v^i)^T F^i = 0, i = 1, \dots, 4 && [F^i \text{ are tangential to the surface}] \\
 & \|F^i\|_2 \leq [\mu[u^i]^T v^i] \nu_i, i = 1, \dots, 4 && [\text{Coulomb's constraints}] \\
 & 0 \leq \nu_i \leq F_{\max}, i = 1, \dots, 4 && [\text{bounds on } \nu_i]
 \end{aligned}$$

in the design variables $\tau, \nu_i, F_i, i = 1, \dots, 4$. This is a conic quadratic program, although not in the standard form (QP). To convert the problem to this standard form, it suffices, e.g., to replace all linear equalities by pairs of linear inequalities and further represent linear inequalities $\alpha^T x \leq \beta$ as conic quadratic constraints

$$Ax - b \equiv \begin{bmatrix} 0 \\ \beta - \alpha^T x \end{bmatrix} \geq_{\mathbf{L}^2} 0.$$

2.3 What can be expressed via conic quadratic constraints?

Optimization problems arising in applications are not normally in their “catalogue” forms, and thus an important skill required from those interested in applications of Optimization is the ability to recognize the fundamental structure underneath the original formulation. The latter is frequently in the form

$$\min_x \{f(x) : x \in X\}, \quad (2.3.1)$$

where f is a “loss function”, and the set X of admissible design vectors is typically given as

$$X = \bigcap_{i=1}^m X_i; \quad (2.3.2)$$

every X_i is the set of vectors admissible for a particular design restriction which in many cases is given by

$$X_i = \{x \in \mathbf{R}^n : g_i(x) \leq 0\}, \quad (2.3.3)$$

where $g_i(x)$ is i -th *constraint function*²⁾.

It is well-known that the objective f in always can be assumed linear, otherwise we could move the original objective to the list of constraints, passing to the equivalent problem

$$\min_{t,x} \left\{ t : (t,x) \in \hat{X} \equiv \{(x,t) : x \in X, t \geq f(x)\} \right\}.$$

Thus, we may assume that the original problem is of the form

$$\min_x \left\{ c^T x : x \in X = \bigcap_{i=1}^m X_i \right\}. \quad (\text{P})$$

²⁾Speaking about a “real-valued function on \mathbf{R}^n ”, we assume that the function is allowed to take real values and the value $+\infty$ and is defined on the entire space. The set of those x where the function is finite is called the domain of the function, denoted by $\text{Dom } f$.

In order to recognize that X is in one of our “catalogue” forms, one needs a kind of dictionary, where different forms of the same structure are listed. We shall build such a dictionary for the conic quadratic programs. Thus, our goal is to understand when a given set X can be represented by *conic quadratic inequalities* (c.q.i.’s), i.e., one or several constraints of the type $\|Dx - d\|_2 \leq p^T x - q$. The word “represented” needs clarification, and here it is:

We say that a set $X \subset \mathbf{R}^n$ can be represented via conic quadratic inequalities (for short: is CQr – Conic Quadratic representable), if there exists a system S of finitely many vector inequalities of the form $A_j \begin{pmatrix} x \\ u \end{pmatrix} - b_j \geq_{\mathbf{L}^{m_j}} 0$ in variables $x \in \mathbf{R}^n$ and additional variables u such that X is the projection of the solution set of S onto the x -space, i.e., $x \in X$ if and only if one can extend x to a solution (x, u) of the system S :

$$x \in X \Leftrightarrow \exists u : A_j \begin{pmatrix} x \\ u \end{pmatrix} - b_j \geq_{\mathbf{L}^{m_j}} 0, \quad j = 1, \dots, N.$$

Every such system S is called a *conic quadratic representation* (for short: a CQR) of the set X ³⁾

The idea behind this definition is clarified by the following observation:

Consider an optimization problem

$$\min_x \{ c^T x : x \in X \}$$

and assume that X is CQr. Then the problem is equivalent to a conic quadratic program. The latter program can be written down explicitly, provided that we are given a CQR of X .

Indeed, let S be a CQR of X , and u be the corresponding vector of additional variables. The problem

$$\min_{x,u} \{ c^T x : (x, u) \text{ satisfy } S \}$$

with design variables x, u is equivalent to the original problem (P), on one hand, and is a conic quadratic program, on the other hand.

Let us call a problem of the form (P) with CQ-representable X a *good* problem.

How to recognize good problems, i.e., how to recognize CQ-representable sets? Well, how we recognize continuity of a given function, like $f(x, y) = \exp\{\sin(x + \exp\{y\})\}$? Normally it is not done by a straightforward verification of the definition of continuity, but by using two kinds of tools:

- A. We know a number of simple functions – a constant, $f(x) = x$, $f(x) = \sin(x)$, $f(x) = \exp\{x\}$, etc. – which indeed are continuous: “once for the entire life” we have verified it directly, by demonstrating that the functions fit the definition of continuity;
- B. We know a number of basic continuity-preserving operations, like taking products, sums, superpositions, etc.

³⁾Note that here we do not impose on the representing system of conic quadratic inequalities S the requirement to satisfy assumption **A**; e.g., the entire space is CQr – it is a solution set of the “system” $|0^T x| \leq 1$ comprised of a single conic quadratic inequality.

When we see that a function is obtained from “simple” functions – those of type A – by operations of type B (as it is the case in the above example), we immediately infer that the function is continuous.

This approach which is common in Mathematics is the one we are about to follow. In fact, we need to answer two kinds of questions:

(?) What are CQ-representable sets

(??) What are CQ-representable functions $g(x)$, i.e., functions which possess CQ-representable epigraphs

$$\text{Epi}\{g\} = \{(x, t) \in \mathbf{R}^n \times \mathbf{R} : g(x) \leq t\}.$$

Our interest in the second question is motivated by the following

Observation: If a function g is CQ-representable, then so are all its level sets $\{x : g(x) \leq a\}$, and every CQ-representation of (the epigraph of) g explicitly induces CQ-representations of the level sets.

Indeed, assume that we have a CQ-representation of the epigraph of g :

$$g(x) \leq t \Leftrightarrow \exists u : \|\alpha_j(x, t, u)\|_2 \leq \beta_j(x, t, u), \quad j = 1, \dots, N,$$

where α_j and β_j are, respectively, vector-valued and scalar *affine* functions of their arguments.

In order to get from this representation a CQ-representation of a level set $\{x : g(x) \leq a\}$, it suffices to fix in the conic quadratic inequalities $\|\alpha_j(x, t, u)\|_2 \leq \beta_j(x, t, u)$ the variable t at the value a .

We list below our “raw materials” – simple functions and sets admitting CQR’s.

2.3.1 Elementary CQ-representable functions/sets

1. A constant function $g(x) \equiv a$.

Indeed, the epigraph of the function $\{(x, t) \mid a \leq t\}$ is given by a linear inequality, and a linear inequality $0 \leq p^T z - q$ is at the same time conic quadratic inequality $\|0\|_2 \leq p^T z - q$.

2. An affine function $g(x) = a^T x + b$.

Indeed, the epigraph of an affine function is given by a linear inequality.

3. The Euclidean norm $g(x) = \|x\|_2$.

Indeed, the epigraph of g is given by the conic quadratic inequality $\|x\|_2 \leq t$ in variables x, t .

4. The squared Euclidean norm $g(x) = x^T x$.

Indeed, $t = \frac{(t+1)^2}{4} - \frac{(t-1)^2}{4}$, so that

$$x^T x \leq t \Leftrightarrow x^T x + \frac{(t-1)^2}{4} \leq \frac{(t+1)^2}{4} \Leftrightarrow \left\| \begin{pmatrix} x \\ \frac{t-1}{2} \end{pmatrix} \right\|_2 \leq \frac{t+1}{2}$$

(check the second \Leftrightarrow !), and the last relation is a conic quadratic inequality.

5. The fractional-quadratic function $g(x, s) = \begin{cases} \frac{x^T x}{s}, & s > 0 \\ 0, & s = 0, x = 0 \\ +\infty, & \text{otherwise} \end{cases}$ (x vector, s scalar).

Indeed, with the convention that $(x^T x)/0$ is 0 or $+\infty$, depending on whether $x = 0$ or not, and taking into account that $ts = \frac{(t+s)^2}{4} - \frac{(t-s)^2}{4}$, we have:

$$\begin{aligned} \left\{ \frac{x^T x}{s} \leq t, s \geq 0 \right\} &\Leftrightarrow \{x^T x \leq ts, t \geq 0, s \geq 0\} \Leftrightarrow \{x^T x + \frac{(t-s)^2}{4} \leq \frac{(t+s)^2}{4}, t \geq 0, s \geq 0\} \\ &\Leftrightarrow \left\| \begin{pmatrix} x \\ \frac{t-s}{2} \end{pmatrix} \right\|_2 \leq \frac{t+s}{2} \end{aligned}$$

(check the third $\Leftrightarrow!$), and the last relation is a conic quadratic inequality.

The level sets of the CQr functions 1 – 5 provide us with a spectrum of “elementary” CQr sets. We add to this spectrum one more set:

6. (A branch of) Hyperbola $\{(t, s) \in \mathbf{R}^2 : ts \geq 1, t > 0\}$.

Indeed,

$$\begin{aligned} \{ts \geq 1, t > 0\} &\Leftrightarrow \left\{ \frac{(t+s)^2}{4} \geq 1 + \frac{(t-s)^2}{4} \& t > 0 \right\} \Leftrightarrow \left\{ \left\| \begin{pmatrix} \frac{t-s}{2} \\ 1 \end{pmatrix} \right\|_2^2 \leq \frac{(t+s)^2}{4} \right\} \\ &\Leftrightarrow \left\{ \left\| \begin{pmatrix} \frac{t-s}{2} \\ 1 \end{pmatrix} \right\|_2 \leq \frac{t+s}{2} \right\} \end{aligned}$$

(check the last $\Leftrightarrow!$), and the latter relation is a conic quadratic inequality.

Next we study simple operations preserving CQ-representability of functions/sets.

2.3.2 Operations preserving CQ-representability of sets

- A. Intersection: If sets $X_i \subset \mathbf{R}^n$, $i = 1, \dots, N$, are CQr, so is their intersection $X = \bigcap_{i=1}^N X_i$.

Indeed, let S_i be CQ-representation of X_i , and u_i be the corresponding vector of additional variables. Then the system S of constraints of the variables (x, u_1, \dots, u_N) :

$$\{(x, u_i) \text{ satisfies } S_i\}, i = 1, \dots, N$$

is a system of conic quadratic inequalities, and this system clearly is a CQ-representation of X .

Corollary 2.3.1 A polyhedral set – a set in \mathbf{R}^n given by finitely many linear inequalities $a_i^T x \leq b_i$, $i = 1, \dots, m$ – is CQr.

Indeed, a polyhedral set is the intersection of finitely many level sets of affine functions, and all these functions (and thus – their level sets) are CQr.

Corollary 2.3.2 If every one of the sets X_i in problem (P) is CQr, then the problem is good – it can be rewritten in the form of a conic quadratic problem, and such a transformation is readily given by CQR’s of the sets X_i , $i = 1, \dots, m$.

Corollary 2.3.3 Adding to a good problem finitely many CQr constraints $x \in X_i$, (e.g., finitely many scalar linear inequalities), we again get a good problem.

- B. Direct product: If sets $X_i \subset \mathbf{R}^{n_i}$, $i = 1, \dots, k$, are CQr, then so is their direct product $X_1 \times \dots \times X_k$.

Indeed, if $S_i = \{\|\alpha_j^i(x_i, u_i)\|_2 \leq \beta_j^i(x_i, u_i)\}_{j=1}^{N_i}$, $i = 1, \dots, k$, are CQR’s of the sets X_i , then the union over i of this system of inequalities, regarded as a system with design variables $x = (x_1, \dots, x_k)$ and additional variables $u = (u_1, \dots, u_k)$ is a CQR for the direct product of X_1, \dots, X_k .

C. Affine image (“Projection”): If a set $X \subset \mathbf{R}^n$ is CQr and $x \mapsto y = \ell(x) = Ax + b$ is an affine mapping of \mathbf{R}^n to \mathbf{R}^k , then the image $\ell(X)$ of the set X under the mapping is CQr.

Indeed, passing to an appropriate bases in \mathbf{R}^n and \mathbf{R}^k , we may assume that the null space of A is comprised of the last $n - p$ vectors of the basis of \mathbf{R}^n , and that the image of A is spanned by the first p vectors of the basis in \mathbf{R}^k . In other words, we may assume that a vector $x \in \mathbf{R}^n$ can be partitioned as $x = \begin{pmatrix} x' \\ x'' \end{pmatrix}$ (x' is p -, and x'' is $(n - p)$ -dimensional), and that a vector $y \in \mathbf{R}^k$ can be partitioned as $y = \begin{pmatrix} y' \\ y'' \end{pmatrix}$ (y' is p -, and y'' is $(k - p)$ -dimensional) in such a way that $A \begin{pmatrix} x' \\ x'' \end{pmatrix} = \begin{pmatrix} Qx' \\ 0 \end{pmatrix}$ with a nonsingular $p \times p$ matrix Q . Thus,

$$\left\{ \begin{pmatrix} y' \\ y'' \end{pmatrix} = A \begin{pmatrix} x' \\ x'' \end{pmatrix} + b \right\} \Leftrightarrow \left\{ x = \begin{pmatrix} Q^{-1}(y' - b') \\ w \end{pmatrix} \text{ for some } w \text{ & } y'' = b'' \right\}.$$

Now let $S = \{\|\alpha_j(x, u)\|_2 \leq \beta_j(x, u)\}_{j=1}^N$ be CQ-representation of X , where u is the corresponding vector of additional variables and α_j, β_j are affine in (x, u) . Then the system of c.q.i.’s in the design variables $y = \begin{pmatrix} y' \\ y'' \end{pmatrix} \in \mathbf{R}^k$ and additional variables $w \in \mathbf{R}^{n-p}$, u :

$$S^+ = \{\|\alpha_j(\begin{pmatrix} Q^{-1}(y' - b') \\ w \end{pmatrix}, u)\|_2 \leq \beta_j(\begin{pmatrix} Q^{-1}(y' - b') \\ w \end{pmatrix}, u)\}_{j=1}^N \& \{\|y'' - b''\|_2 \leq 0\}$$

is a CQR of $\ell(X)$. Indeed, $y = \begin{pmatrix} y' \\ y'' \end{pmatrix} \in \ell(X)$ if and only if $y'' = b''$ and there exists $w \in \mathbf{R}^{n-p}$ such that the point $x = \begin{pmatrix} Q^{-1}(y' - b') \\ w \end{pmatrix}$ belongs to X , and the latter happens if and only if there exist u such that the point $(x, u) = (\begin{pmatrix} Q^{-1}(y' - b') \\ w \end{pmatrix}, u)$ solves S .

Corollary 2.3.4 *A nonempty set X is CQr if and only if its characteristic function*

$$\chi(x) = \begin{cases} 0, & x \in X \\ +\infty, & \text{otherwise} \end{cases}$$

is CQr.

Indeed, $\text{Epi}\{\chi\}$ is the direct product of X and the nonnegative ray; therefore if X is CQr, so is $\chi(\cdot)$ (see B. and Corollary 2.3.1). Vice versa, if χ is CQr, then X is CQr by C., since X is the projection of the $\text{Epi}\{\chi\}$ on the space of x -variables.

D. Inverse affine image: Let $X \subset \mathbf{R}^n$ be a CQr set, and let $\ell(y) = Ay + b$ be an affine mapping from \mathbf{R}^k to \mathbf{R}^n . Then the inverse image $\ell^{-1}(X) = \{y \in \mathbf{R}^k : Ay + b \in X\}$ of X under the mapping is also CQr.

Indeed, let $S = \{\|\alpha_j(x, u)\|_2 \leq \beta_j(x, u)\}_{j=1}^N$ be a CQR for X . Then the system of c.q.i.’s

$$S = \{\|\alpha_j(Ay + b, u)\|_2 \leq \beta_j(Ay + b, u)\}_{j=1}^N$$

with variables y, u clearly is a CQR for $\ell^{-1}(X)$.

Corollary 2.3.5 *Consider a good problem (P) and assume that we restrict its design variables to be given affine functions of a new design vector y . Then the induced problem with the design vector y is also good.*

In particular, adding to a good problem arbitrarily many linear equality constraints, we end up with a good problem (Indeed, we may use the linear equations to express affinely the original design variables via part of them, let this part be y ; the problem with added linear constraints can now be posed as a problem with design vector y).

It should be stressed that the above statements are not just existence theorems – they are “algorithmic”: given CQR’s of the “operands” (say, m sets X_1, \dots, X_m), we may build *completely mechanically* a CQR for the “result of the operation” (e.g., for the intersection $\bigcap_{i=1}^m X_i$).

2.3.3 Operations preserving CQ-representability of functions

Recall that a function $g(x)$ is called CQ-representable, if its epigraph $\text{Epi}\{g\} = \{(x, t) : g(x) \leq t\}$ is a CQ-representable set; a CQR of the epigraph of g is called conic quadratic representation of g . Recall also that a level set of a CQR function is CQ-representable. Here are transformations preserving CQ-representability of functions:

E. Taking maximum: If functions $g_i(x)$, $i = 1, \dots, m$, are CQR, then so is their maximum $g(x) = \max_{i=1, \dots, m} g_i(x)$.

Indeed, $\text{Epi}\{g\} = \bigcap_i \text{Epi}\{g_i\}$, and the intersection of finitely many CQR sets again is CQR.

F. Summation with nonnegative weights: If functions $g_i(x)$, $x \in \mathbf{R}^n$, are CQR, $i = 1, \dots, m$, and α_i are nonnegative weights, then the function $g(x) = \sum_{i=1}^m \alpha_i g_i(x)$ is also CQR.

Indeed, consider the set

$$\Pi = \{(x_1, t_1; x_2, t_2; \dots; x_m, t_m; t) : x_i \in \mathbf{R}^n, t_i \in \mathbf{R}, t \in \mathbf{R}, g_i(x_i) \leq t_i, i = 1, \dots, m; \sum_{i=1}^m \alpha_i t_i \leq t\}.$$

The set is CQR. Indeed, the set is the direct product of the epigraphs of g_i intersected with the half-space given by the linear inequality $\sum_{i=1}^m \alpha_i t_i \leq t$. Now, a direct product of CQR sets is also CQR, a half-space is CQR (it is a level set of an affine function, and such a function is CQR), and the intersection of CQR sets is also CQR. Since Π is CQR, so is its projection on subspace of variables x_1, x_2, \dots, x_m, t , i.e., the set

$$\{(x_1, \dots, x_m, t) : \exists t_1, \dots, t_m : g_i(x_i) \leq t_i, i = 1, \dots, m, \sum_{i=1}^m \alpha_i t_i \leq t\} = \{(x_1, \dots, x_m, t) : \sum_{i=1}^m \alpha_i g_i(x) \leq t\}.$$

Since the latter set is CQR, so is its inverse image under the mapping

$$(x, t) \mapsto (x, x, \dots, x, t),$$

and this inverse image is exactly the epigraph of g .

G. Direct summation: If functions $g_i(x_i)$, $x_i \in \mathbf{R}^{n_i}$, $i = 1, \dots, m$, are CQR, so is their direct sum

$$g(x_1, \dots, x_m) = g_1(x_1) + \dots + g_m(x_m).$$

Indeed, the functions $\hat{g}_i(x_1, \dots, x_m) = g_i(x_i)$ are clearly CQR – their epigraphs are inverse images of the epigraphs of g_i under the affine mappings $(x_1, \dots, x_m, t) \mapsto (x_i, t)$. It remains to note that $g = \sum_i \hat{g}_i$.

H. Affine substitution of argument: If a function $g(x)$, $x \in \mathbf{R}^n$, is CQR and $y \mapsto Ay + b$ is an affine mapping from \mathbf{R}^k to \mathbf{R}^n , then the superposition $g^\rightarrow(y) = g(Ay + b)$ is CQR.

Indeed, the epigraph of g^\rightarrow is the inverse image of the epigraph of g under the affine mapping $(y, t) \mapsto (Ay + b, t)$.

I. Partial minimization: Let $g(x)$ be CQr. Assume that x is partitioned into two sub-vectors: $x = (v, w)$, and let \hat{g} be obtained from g by partial minimization in w :

$$\hat{g}(v) = \inf_w g(v, w),$$

and assume that for every v the minimum in w is achieved. Then \hat{g} is CQr.

Indeed, under the assumption that the minimum in w always is achieved, $\text{Epi}\{\hat{g}\}$ is the image of the epigraph of $\text{Epi}\{g\}$ under the projection $(v, w, t) \mapsto (v, t)$.

2.3.4 More operations preserving CQ-representability

Let us list a number of more “advanced” operations with sets/functions preserving CQ-representability.

J. Arithmetic summation of sets. Let X_i , $i = 1, \dots, k$, be nonempty convex sets in \mathbf{R}^n , and let $X_1 + X_2 + \dots + X_k$ be the arithmetic sum of these sets:

$$X_1 + \dots + X_k = \{x = x^1 + \dots + x^k : x^i \in X_i, i = 1, \dots, k\}.$$

We claim that

If all X_i are CQr, so is their sum.

Indeed, the direct product

$$X = X_1 \times X_2 \times \dots \times X_k \subset \mathbf{R}^{nk}$$

is CQr by B.; it remains to note that $X_1 + \dots + X_k$ is the image of X under the linear mapping

$$(x^1, \dots, x^k) \mapsto x^1 + \dots + x^k : \mathbf{R}^{nk} \rightarrow \mathbf{R}^n,$$

and by C. the image of a CQr set under an affine mapping is also CQr (see C.)

J.1. inf-convolution. The operation with functions related to the arithmetic summation of sets is the inf-convolution defined as follows. Let $f_i : \mathbf{R}^n \rightarrow \mathbf{R} \cup \{\infty\}$, $i = 1, \dots, n$, be functions. Their inf-convolution is the function

$$f(x) = \inf\{f_1(x^1) + \dots + f_k(x^k) : x^1 + \dots + x^k = x\}. \quad (*)$$

We claim that

If all f_i are CQr, their inf-convolution is $> -\infty$ everywhere and for every x for which the inf in the right hand side of $()$ is finite, this infimum is achieved, then f is CQr.*

Indeed, under the assumption in question the epigraph $\text{Epi}\{f\} = \text{Epi}\{f_1\} + \dots + \text{Epi}\{f_k\}$.

K. Taking conic hull of a convex set. Let $X \in \mathbf{R}^n$ be a nonempty convex set. Its *conic hull* is the set

$$X^+ = \{(x, t) \in \mathbf{R}^n \times \mathbf{R} : t > 0, t^{-1}x \in X\}.$$

Geometrically: we add to the coordinates of vectors from \mathbf{R}^n a new coordinate equal to 1:

$$(x_1, \dots, x_n)^T \mapsto (x_1, \dots, x_n, 1)^T,$$

thus getting an affine embedding of \mathbf{R}^n in \mathbf{R}^{n+1} . We take the image of X under this mapping – “lift” X by one along the $(n+1)$ st axis – and then form the set X^+ by taking all (open) rays emanating from the origin and crossing the “lifted” X .

The conic hull is not closed (e.g., it does not contain the origin, which clearly is in its closure). The closed convex hull of X is the closure of its conic hull:

$$\widehat{X}^+ = \text{cl } X^+ = \left\{ (x, t) \in \mathbf{R}^n \times \mathbf{R} : \exists \{(x_i, t_i)\}_{i=1}^\infty : t_i > 0, t_i^{-1}x_i \in X, t = \lim_i t_i, x = \lim_i x_i \right\}.$$

Note that if X is a closed convex set, then the conic hull X^+ of X is nothing but the intersection of the closed convex hull \widehat{X}^+ and the open half-space $\{t > 0\}$ (check!); thus, the closed conic hull of a closed convex set X is larger than the conic hull by some part of the hyperplane $\{\tau = 0\}$. When X is closed and bounded, then the difference between the hulls is pretty small: $\widehat{X}^+ = X^+ \cup \{0\}$ (check!). Note also that if X is a closed convex set, you can obtain it from its (closed) convex hull by taking intersection with the hyperplane $\{t = 1\}$:

$$x \in X \Leftrightarrow (x, 1) \in \widehat{X}^+ \Leftrightarrow (x, 1) \in X^+.$$

Proposition 2.3.1 (i) If a set X is CQr:

$$X = \{x : \exists u : Ax + Bu + b \geq_{\mathbf{K}} 0\}, \quad (2.3.4)$$

where \mathbf{K} is a direct product of the ice-cream cones, then the conic hull X^+ is CQr as well:

$$X^+ = \{(x, t) : \exists (u, s) : Ax + Bu + tb \geq_{\mathbf{K}} 0, \begin{bmatrix} 2 \\ s-t \\ s+t \end{bmatrix} \geq_{\mathbf{L}^3} 0\}. \quad (2.3.5)$$

(ii) If the set X given by (2.3.4) is closed, then the CQr set

$$\tilde{X}^+ = \{(x, t) : \exists u : Ax + Bu + tb \geq_{\mathbf{K}} 0\} \cap \{(x, t) : t \geq 0\} \quad (2.3.6)$$

is “between” the conic hull X^+ and the closed conic hull \widehat{X}^+ of X :

$$X^+ \subset \tilde{X}^+ \subset \widehat{X}^+.$$

(iii) If the CQR (2.3.4) is such that $Bu \in \mathbf{K}$ implies that $Bu = 0$, then $\tilde{X}^+ = \widehat{X}^+$, so that \widehat{X}^+ is CQr.

(i): We have

$$\begin{aligned} X^+ &\equiv \{(x, t) : t > 0, x/t \in X\} \\ &= \{(x, t) : \exists u : A(x/t) + Bu + b \geq_{\mathbf{K}} 0, t > 0\} \\ &= \{(x, t) : \exists v : Ax + Bv + tb \geq_{\mathbf{K}} 0, t > 0\} \\ &= \{(x, t) : \exists v, s : Ax + Bv + tb \geq_{\mathbf{K}} 0, t, s \geq 0, ts \geq 1\}, \end{aligned}$$

and we arrive at (2.3.5).

(ii): We should prove that the set \tilde{X}^+ (which by construction is CQr) is between X^+ and \widehat{X}^+ . The inclusion $X^+ \subset \tilde{X}^+$ is readily given by (2.3.5). Next, let us prove that $\tilde{X}^+ \subset \widehat{X}^+$. Let us choose a point $\bar{x} \in X$, so that for a properly chosen \bar{u} it holds

$$A\bar{x} + B\bar{u} + b \geq_{\mathbf{K}} 0,$$

i.e., $(\bar{x}, 1) \in \tilde{X}^+$. Since \tilde{X}^+ is convex (this is true for every CQr set), we conclude that whenever (x, t) belongs to \tilde{X}^+ , so does every pair $(x_\epsilon = x + \epsilon\bar{x}, t_\epsilon = t + \epsilon)$ with $\epsilon > 0$:

$$\exists u = u_\epsilon : Ax_\epsilon + Bu_\epsilon + t_\epsilon b \geq_{\mathbf{K}} 0.$$

It follows that $t_\epsilon^{-1}x_\epsilon \in X$, whence $(x_\epsilon, t_\epsilon) \in X^+ \subset \widehat{X}^+$. As $\epsilon \rightarrow +0$, we have $(x_\epsilon, t_\epsilon) \rightarrow (x, t)$, and since \widehat{X}^+ is closed, we get $(x, t) \in \widehat{X}^+$. Thus, $\tilde{X}^+ \subset \widehat{X}^+$.

(ii): Assume that $Bu \in \mathbf{K}$ only if $Bu = 0$, and let us show that $\tilde{X}^+ = \widehat{X}^+$. We just have to prove that \tilde{X}^+ is closed, which indeed is the case due to the following

Lemma 2.3.1 Let Y be a CQr set with CQR

$$Y = \{y : \exists v : Py + Qv + r \geq_{\mathbf{K}} 0\}$$

such that $Qv \in \mathbf{K}$ only when $Qv = 0$. Then

- (i) There exists a constant $C < \infty$ such that

$$Py + Qv + r \in \mathbf{K} \Rightarrow \|Qv\|_2 \leq C(1 + \|Py + r\|_2); \quad (2.3.7)$$

- (ii) Y is closed.

Proof of Lemma. (i): Assume, on the contrary to what should be proved, that there exists a sequence $\{y_i, v_i\}$ such that

$$Py_i + Qv_i + r \in \mathbf{K}, \quad \|Qv_i\|_2 \geq \alpha_i(1 + \|Py_i + r\|_2), \quad i \rightarrow \infty \text{ as } i \rightarrow \infty. \quad (2.3.8)$$

By Linear Algebra, for every b such that the linear system $Qv = b$ is solvable, it admits a solution v such that $\|v\|_2 \leq C_1\|b\|_2$ with $C_1 < \infty$ depending on Q only; therefore we can assume, in addition to (2.3.8), that

$$\|v_i\|_2 \leq C_1\|Qv_i\|_2 \quad (2.3.9)$$

for all i . Now, from (2.3.8) it clearly follows that

$$\|Qv_i\|_2 \rightarrow \infty \text{ as } i \rightarrow \infty; \quad (2.3.10)$$

setting

$$\widehat{v}_i = \frac{1}{\|Qv_i\|_2}v_i,$$

we have

- (a) $\|Q\widehat{v}_i\|_2 = 1 \quad \forall i,$
- (b) $\|\widehat{v}_i\| \leq C_1 \quad \forall i, \quad [\text{by (2.3.9)}]$
- (c) $Q\widehat{v}_i + \|Qv_i\|_2^{-1}(Py_i + r) \in \mathbf{K} \quad \forall i,$
- (d) $\|Qv_i\|_2^{-1}\|Py_i + r\|_2 \leq \alpha_i^{-1} \rightarrow 0 \text{ as } i \rightarrow \infty \quad [\text{by (2.3.8)}]$

Taking into account (b) and passing to a subsequence, we can assume that $\widehat{v}_i \rightarrow \widehat{v}$ as $i \rightarrow \infty$; by (c, d) $Q\widehat{v} \in \mathbf{K}$, while by (a) $\|Q\widehat{v}\|_2 = 1$, i.e., $Q\widehat{v} \neq 0$, which is the desired contradiction.

(ii) To prove that Y is closed, assume that $y_i \in Y$ and $y_i \rightarrow y$ as $i \rightarrow \infty$, and let us verify that $y \in Y$. Indeed, since $y_i \in Y$, there exist v_i such that $Py_i + Qv_i + r \in \mathbf{K}$. Same as above, we can assume that (2.3.9) holds. Since $y_i \rightarrow y$, the sequence $\{Qv_i\}$ is bounded by (2.3.7), so that the sequence $\{v_i\}$ is bounded by (2.3.9). Passing to a subsequence, we can assume that $v_i \rightarrow v$ as $i \rightarrow \infty$; passing to the limit, as $i \rightarrow \infty$, in the inclusion $Py_i + Qv_i + r \in \mathbf{K}$, we get $Py + Qv + r \in \mathbf{K}$, i.e., $y \in Y$. ■

K.1. “Projective transformation” of a CQr function. The operation with functions related to taking conic hull of a convex set is the “projective transformation” which converts a function $f(x) : \mathbf{R}^n \rightarrow \mathbf{R} \cup \{\infty\}$ ⁴⁾ into the function

$$f^+(x, s) = sf(x/s) : \{s > 0\} \times \mathbf{R}^n \rightarrow \mathbf{R} \cup \{\infty\}.$$

The epigraph of f^+ is the conic hull of the epigraph of f with the origin excluded:

$$\begin{aligned} \{(x, s, t) : s > 0, t \geq f^+(x, s)\} &= \{(x, s, t) : s > 0, s^{-1}t \geq f(s^{-1}x)\} \\ &= \{(x, s, t) : s > 0, s^{-1}(x, t) \in \text{Epi}\{f\}\}. \end{aligned}$$

The closure $\text{cl Epi}\{f^+\}$ is the epigraph of certain function, let it be denoted $\widehat{f}^+(x, s)$; this function is called the *projective transformation* of f . E.g., the fractional-quadratic function from Example 5 is

⁴⁾ Recall that “a function” for us means a *proper* function – one which takes a finite value at least at one point

the projective transformation of the function $f(x) = x^T x$. Note that the function $\widehat{f}^+(x, s)$ does not necessarily coincide with $f^+(x, s)$ even in the open half-space $s > 0$; this is the case if and only if the epigraph of f is closed (or, which is the same, f is *lower semicontinuous*: whenever $x_i \rightarrow x$ and $f(x_i) \rightarrow a$, we have $f(x) \leq a$). We are about to demonstrate that the projective transformation “nearly” preserves CQ-representability:

Proposition 2.3.2 *Let $f : \mathbf{R}^n \rightarrow \mathbf{R} \cup \{\infty\}$ be a lower semicontinuous function which is CQr:*

$$\text{Epi}\{f\} \equiv \{(x, t) : t \geq f(x)\} = \{(t, x) : \exists u : Ax + tp + Bu + b \geq_{\mathbf{K}} 0\}, \quad (2.3.11)$$

where \mathbf{K} is a direct product of ice-cream cones. Assume that the CQR is such that $Bu \geq_{\mathbf{K}} 0$ implies that $Bu = 0$. Then the projective transformation \widehat{f}^+ of f is CQr, namely,

$$\text{Epi}\{\widehat{f}^+\} = \{(x, t, s) : s \geq 0, \exists u : Ax + tp + Bu + sb \geq_{\mathbf{K}} 0\}.$$

Indeed, let us set

$$G = \{(x, t, s) : \exists u : s \geq 0, Ax + tp + Bu + sb \geq_{\mathbf{K}} 0\}.$$

As we remember from the previous combination rule, G is exactly the closed conic hull of the epigraph of f , i.e., $G = \text{Epi}\{\widehat{f}^+\}$.

L. The polar of a convex set. Let $X \subset \mathbf{R}^n$ be a convex set containing the origin. The *polar* of X is the set

$$X_* = \{y \in \mathbf{R}^n : y^T x \leq 1 \ \forall x \in X\}.$$

In particular,

- the polar of the singleton $\{0\}$ is the entire space;
- the polar of the entire space is the singleton $\{0\}$;
- the polar of a linear subspace is its orthogonal complement (why?);
- the polar of a closed convex pointed cone K with a nonempty interior is $-K_*$, minus the dual cone (why?).

Polarity is “symmetric”: if X is a closed convex set containing the origin, then so is X_* , and twice taken polar is the original set: $(X_*)_* = X$.

We are about to prove that the polarity $X \mapsto X_*$ “nearly” preserves CQ-representability:

Proposition 2.3.3 *Let $X \subset \mathbf{R}^n$, $0 \in X$, be a CQr set:*

$$X = \{x : \exists u : Ax + Bu + b \geq_{\mathbf{K}} 0\}, \quad (2.3.12)$$

where \mathbf{K} is a direct product of ice-cream cones.

Assume that there exists \bar{x}, \bar{u} such that

$$A\bar{x} + B\bar{u} + b >_{\mathbf{K}} 0.$$

Then the polar of X is the CQr set

$$X_* = \{y : \exists \xi : A^T \xi + y = 0, B^T \xi = 0, b^T \xi \leq 1, \xi \geq_{\mathbf{K}} 0\} \quad (2.3.13)$$

Indeed, consider the following conic quadratic problem:

$$\min_{x, u} \{-y^T x : Ax + Bu + b \geq_{\mathbf{K}} 0\}. \quad (\mathcal{P}_y)$$

A vector y belongs to X_* if and only if (P_y) is bounded below and its optimal value is at least -1 . Since (P_y) is strictly feasible, from the Conic Duality Theorem it follows that these properties of (P_y) hold if and only if the dual problem

$$-b^T \xi \rightarrow \max : A^T \xi = -y, B^T \xi = 0, \xi \geq_{\mathbf{K}} 0$$

(recall that \mathbf{K} is self-dual) has a feasible solution with the value of the dual objective at least -1 . Thus,

$$X_* = \{y : \exists \xi : A^T \xi + y = 0, B^T \xi = 0, b^T \xi \leq 1, \xi \geq_{\mathbf{K}} 0\},$$

as claimed in (2.3.13). It remains to note that X_* is obtained from the CQr set \mathbf{K} by operations preserving CQ-representability: intersection with the CQr set $\{\xi : B^T \xi = 0, b^T \xi \leq 1\}$ and subsequent affine mapping $\xi \mapsto -A^T \xi$.

L.1. The Legendre transformation of a CQr function. The operation with functions related to taking polar of a convex set is the *Legendre* (or *conjugate*) *transformation*. The Legendre transformation (\equiv the conjugate) of a function $f(x) : \mathbf{R}^n \rightarrow \mathbf{R} \cup \{\infty\}$ is the function

$$f_*(y) = \sup_x [y^T x - f(x)].$$

In particular,

- the conjugate of a constant $f(x) \equiv c$ is the function

$$f_*(y) = \begin{cases} -c, & y = 0 \\ +\infty, & y \neq 0 \end{cases};$$

- the conjugate of an affine function $f(x) \equiv a^T x + b$ is the function

$$f_*(y) = \begin{cases} -b, & y = a \\ +\infty, & y \neq a \end{cases};$$

- the conjugate of a convex quadratic form $f(x) \equiv \frac{1}{2}x^T D^T D x + b^T x + c$ with rectangular D such that $\text{Null}(D^T) = \{0\}$ is the function

$$f_*(y) = \begin{cases} \frac{1}{2}(y - b)^T D^T (DD^T)^{-2} D(y - b) - c, & y - b \in \text{Im}D^T \\ +\infty, & \text{otherwise} \end{cases};$$

It is worth mentioning that the Legendre transformation is symmetric: if f is a proper convex lower semicontinuous function (i.e., $\emptyset \neq \text{Epi}\{f\}$ is convex and closed), then so is f_* , and taken twice, the Legendre transformation recovers the original function: $(f_*)_* = f$.

We are about to prove that the Legendre transformation “nearly” preserves CQ-representability:

Proposition 2.3.4 Let $f : \mathbf{R}^n \rightarrow \mathbf{R} \cup \{\infty\}$ be CQr:

$$\{(x, t) : t \geq f(x)\} = \{(t, x) : \exists u : Ax + tp + Bu + b \geq_{\mathbf{K}} 0\},$$

where \mathbf{K} is a direct product of ice-cream cones. Assume that there exist $\bar{x}, \bar{t}, \bar{u}$ such that

$$A\bar{x} + \bar{t}p + B\bar{u} + b >_{\mathbf{K}} 0.$$

Then the Legendre transformation of f is CQr:

$$\text{Epi}\{f_*\} = \{(y, s) : \exists \xi : A^T \xi = -y, B^T \xi = 0, p^T \xi = 1, s \geq b^T \xi, \xi \geq_{\mathbf{K}} 0\}. \quad (2.3.14)$$

Indeed, we have

$$\text{Epi}\{f_*\} = \{(y, s) : y^T x - f(x) \leq s \ \forall x\} = \{(y, s) : y^T x - t \leq s \ \forall (x, t) \in \text{Epi}\{f\}\}. \quad (2.3.15)$$

Consider the conic quadratic program

$$\min_{x, t, u} \{-y^T x + t : Ax + tp + Bu + b \geq \mathbf{K}_0\}. \quad (\text{P}_y)$$

By (2.3.15), a pair (y, s) belongs to $\text{Epi}\{f_*\}$ if and only if (P_y) is bounded below with optimal value $\geq -s$. Since (P_y) is strictly feasible, this is the case if and only if the dual problem

$$\max_{\xi} \{-b^T \xi : A^T \xi = -y, B^T \xi = 0, p^T \xi = 1, \xi \geq_{\mathbf{K}} 0\}$$

has a feasible solution with the value of the dual objective $\geq -s$. Thus,

$$\text{Epi}\{f_*\} = \{(y, s) : \exists \xi : A^T \xi = -y, B^T \xi = 0, p^T \xi = 1, s \geq b^T \xi, \xi \geq_{\mathbf{K}} 0\}$$

as claimed in (2.3.14). It remains to note that the right hand side set in (2.3.14) is CQr (as a set obtained from the CQr set $\mathbf{K} \times \mathbf{R}_s$ by operations preserving CQ-representability – intersection with the set $\{\xi : B^T \xi = 0, p^T \xi = 1, b^T \xi \leq s\}$ and subsequent affine mapping $\xi \mapsto -A^T \xi$).

Corollary 2.3.6 *The support function*

$$\text{Supp}_X(x) = \sup_{y \in X} x^T y$$

of a nonempty CQr set X is CQr.

Indeed, $\text{Supp}_X(\cdot)$ is the conjugate of the characteristic function of X , and it remains to refer to Corollary 2.3.4.

M. Taking convex hull of several sets. The convex hull of a set $Y \subset \mathbf{R}^n$ is the smallest convex set which contains Y :

$$\text{Conv}(Y) = \left\{ x = \sum_{i=1}^{k_x} \alpha_i x_i : x_i \in Y, \alpha_i \geq 0, \sum_i \alpha_i = 1 \right\}$$

The closed convex hull $\overline{\text{Conv}}(Y) = \text{cl Conv}(Y)$ of Y is the smallest closed convex set containing Y .

Following Yu. Nesterov, let us prove that taking convex hull “nearly” preserves CQ-representability:

Proposition 2.3.5 *Let $X_1, \dots, X_k \subset \mathbf{R}^n$ be closed convex CQr sets:*

$$X_i = \{x : A_i x + B_i u_i + b_i \geq_{\mathbf{K}_i} 0, i = 1, \dots, k\}, \quad (2.3.16)$$

where \mathbf{K}_i is a direct product of ice-cream cones.

Then the CQr set

$$\begin{aligned} Y &= \left\{ x : \begin{array}{l} \exists \xi^1, \dots, \xi^k, t_1, \dots, t_k, \eta^1, \dots, \eta^k : \\ \begin{bmatrix} A_1 \xi^1 + B_1 \eta^1 + t_1 b_1 \\ A_2 \xi^2 + B_2 \eta^2 + t_2 b_2 \\ \vdots \\ A_k \xi^k + B_k \eta^k + t_k b_k \end{bmatrix} \geq_{\mathbf{K}} 0, \\ t_1, \dots, t_k \geq 0, \\ \xi^1 + \dots + \xi^k = x \\ t_1 + \dots + t_k = 1 \end{array} \right\}, \\ \mathbf{K} &= \mathbf{K}_1 \times \dots \times \mathbf{K}_k \end{aligned} \quad (2.3.17)$$

is between the convex hull and the closed convex hull of the set $X_1 \cup \dots \cup X_k$:

$$\text{Conv}(\bigcup_{i=1}^k X_i) \subset Y \subset \overline{\text{Conv}}(\bigcup_{i=1}^k X_i).$$

If, in addition to CQ-representability,

- (i) all X_i are bounded,

or

- (ii) $X_i = Z_i + W$, where Z_i are closed and bounded sets and W is a convex closed set, then

$$\text{Conv}(\bigcup_{i=1}^k X_i) = Y = \overline{\text{Conv}}(\bigcup_{i=1}^k X_i)$$

is CQr.

First, the set Y clearly contains $\text{Conv}(\bigcup_{i=1}^k X_i)$. Indeed, since the sets X_i are convex, the convex hull of their union is

$$\left\{ x = \sum_{i=1}^k t_i x^i : x^i \in X_i, t_i \geq 0, \sum_{i=1}^k t_i = 1 \right\}$$

(why?); for a point

$$x = \sum_{i=1}^k t_i x^i \quad \left[x^i \in X_i, t_i \geq 0, \sum_{i=1}^k t_i = 1 \right],$$

there exist u^i , $i = 1, \dots, k$, such that

$$A_i x^i + B_i u^i + b_i \geq_{\mathbf{K}_i} 0.$$

We get

$$\begin{aligned} x &= (t_1 x^1) + \dots + (t_k x^k) \\ &= \xi^1 + \dots + \xi^k, \\ &\quad [\xi^i = t_i x^i]; \\ t_1, \dots, t_k &\geq 0; \\ t_1 + \dots + t_k &= 1; \\ A_i \xi^i + B_i \eta^i + t_i b_i &\geq_{\mathbf{K}_i} 0, \quad i = 1, \dots, k, \\ &\quad [\eta^i = t_i u^i], \end{aligned} \tag{2.3.18}$$

so that $x \in Y$ (see the definition of Y).

To complete the proof that Y is between the convex hull and the closed convex hull of $\bigcup_{i=1}^k X_i$, it remains to verify that if $x \in Y$ then x is contained in the closed convex hull of $\bigcup_{i=1}^k X_i$. Let us somehow choose $\bar{x}^i \in X_i$; for properly chosen \bar{u}^i we have

$$A_i \bar{x}^i + B_i \bar{u}^i + b_i \geq_{\mathbf{K}_i} 0, \quad i = 1, \dots, k. \tag{2.3.19}$$

Since $x \in Y$, there exist t_i, ξ^i, η^i satisfying the relations

$$\begin{aligned} x &= \xi^1 + \dots + \xi^k, \\ t_1, \dots, t_k &\geq 0, \\ t_1 + \dots + t_k &= 1, \\ A_i \xi^i + B_i \eta^i + t_i b_i &\geq_{\mathbf{K}_i} 0, \quad i = 1, \dots, k. \end{aligned} \tag{2.3.20}$$

In view of the latter relations and (2.3.19), we have for $0 < \epsilon < 1$:

$$A_i[(1-\epsilon)\xi^i + \epsilon k^{-1}\bar{x}^i] + B_i[(1-\epsilon)\eta^i + \epsilon k^{-1}\bar{u}^i] + [(1-\epsilon)t_i + \epsilon k^{-1}]b_i \geq_{\mathbf{K}_i} 0;$$

setting

$$\begin{aligned} t_{i,\epsilon} &= (1-\epsilon)t_i + \epsilon k^{-1}; \\ x_\epsilon^i &= t_{i,\epsilon}^{-1}[(1-\epsilon)\xi^i + \epsilon k^{-1}\bar{x}^i]; \\ u_\epsilon^i &= t_{i,\epsilon}^{-1}[(1-\epsilon)\eta^i + \epsilon k^{-1}\bar{u}^i], \end{aligned}$$

we get

$$\begin{aligned} A_i x_\epsilon^i + B_i u_\epsilon^i + b_i &\geq_{\mathbf{K}_i} 0 \Rightarrow \\ x_\epsilon^i &\in X_i, \\ t_{1,\epsilon}, \dots, t_{k,\epsilon} &\geq 0, \\ t_{1,\epsilon} + \dots + t_{k,\epsilon} &= 1 \\ \Rightarrow x_\epsilon &\equiv \sum_{i=1}^k t_{i,\epsilon} x_\epsilon^i \\ &\in \text{Conv}(\bigcup_{i=1}^k X_i). \end{aligned}$$

On the other hand, we have by construction

$$x_\epsilon = \sum_{i=1}^k [(1-\epsilon)\xi^i + \epsilon k^{-1}\bar{x}^i] \rightarrow x = \sum_{i=1}^k \xi^i \text{ as } \epsilon \rightarrow +0,$$

so that x belongs to the closed convex hull of $\bigcup_{i=1}^k X_i$, as claimed.

It remains to verify that in the cases of (i), (ii) the convex hull of $\bigcup_{i=1}^k X_i$ is the same as the closed convex hull of this union. (i) is a particular case of (ii) corresponding to $W = \{0\}$, so that it suffices to prove (ii). Assume that

$$\begin{aligned} x_t &= \sum_{i=1}^k \mu_{ti}[z_{ti} + p_{ti}] \rightarrow x \text{ as } t \rightarrow \infty \\ \left[z_{ti} \in Z_i, p_{ti} \in W, \mu_{ti} \geq 0, \sum_i \mu_{ti} = 1 \right] \end{aligned}$$

and let us prove that x belongs to the convex hull of the union of X_i . Indeed, since Z_i are closed and bounded, passing to a subsequence, we may assume that

$$z_{ti} \rightarrow z_i \in Z_i \text{ and } \mu_{ti} \rightarrow \mu_i \text{ as } t \rightarrow \infty.$$

It follows that the vectors

$$p_t = \sum_{i=1}^m \mu_{ti} p_{ti} = x_t - \sum_{i=1}^k \mu_{ti} z_{ti}$$

converge as $t \rightarrow \infty$ to some vector p , and since W is closed and convex, $p \in W$. We now have

$$x = \lim_{t \rightarrow \infty} \left[\sum_{i=1}^k \mu_{ti} z_{ti} + p_t \right] = \sum_{i=1}^k \mu_i z_i + p = \sum_{i=1}^k \mu_i [z_i + p],$$

so that x belongs to the convex hull of the union of X_i (as a convex combination of points $z_i + p \in X_i$).

P. The recessive cone of a CQr set. Let X be a closed convex set. The recessive cone $\text{Rec}(X)$ of X is the set

$$\text{Rec}(X) = \{h : x + th \in X \quad \forall(x \in X, t \geq 0)\}.$$

It can be easily verified that $\text{Rec}(X)$ is a closed cone, and that

$$\text{Rec}(X) = \{h : \bar{x} + th \in X \quad \forall t \geq 0\} \quad \forall \bar{x} \in X,$$

i.e., that $\text{Rec}(X)$ is the set of all directions h such that the ray emanating from a point of X and directed by h is contained in X .

Proposition 2.3.6 *Let X be a nonempty CQr set with CQR*

$$X = \{x \in \mathbf{R}^n : \exists u : Ax + Bu + b \geq_{\mathbf{K}} 0\},$$

where \mathbf{K} is a direct product of ice-cream cones, and let the CQR be such that $Bu \in \mathbf{K}$ only if $Bu = 0$. Then X is closed, and the recessive cone of X is CQr:

$$\text{Rec}(X) = \{h : \exists v : Ah + Bv \geq_{\mathbf{K}} 0\}. \quad (2.3.21)$$

Proof. The fact that X is closed is given by Lemma 2.3.1. In order to prove (2.3.21), let us temporary denote by R the set in the left hand side of this relation; we should prove that $R = \text{Rec}(X)$. The inclusion $R \subset \text{Rec}(X)$ is evident. To prove the inverse inclusion, let $\bar{x} \in X$ and $h \in \text{Rec}(X)$, so that for every $i = 1, 2, \dots$ there exists u_i such that

$$A(\bar{x} + ih) + Bu_i + b \in \mathbf{K}. \quad (2.3.22)$$

By Lemma 2.3.1,

$$\|Bu_i\|_2 \leq C(1 + \|A(\bar{x} + ih) + b\|_2) \quad (2.3.23)$$

for certain $C < \infty$ and all i . Besides this, we can assume w.l.o.g. that

$$\|u_i\|_2 \leq C_1\|Bu_i\|_2 \quad (2.3.24)$$

(cf. the proof of Lemma 2.3.1). By (2.3.23) – (2.3.24), the sequence $\{v_i = i^{-1}u_i\}$ is bounded; passing to a subsequence, we can assume that $v_i \rightarrow v$ as $i \rightarrow \infty$. By (2.3.22), we have for all i

$$i^{-1}A(\bar{x} + ih) + Bv_i + i^{-1}b \in \mathbf{K},$$

whence, passing to limit as $i \rightarrow \infty$, $Ah + Bv \in \mathbf{K}$. Thus, $h \in R$. ■

O. Theorem on superposition. Let $f_\ell : \mathbf{R}^m \rightarrow \mathbf{R} \cup \{+\infty\}$, $\ell = 1, \dots, m$ be CQr functions:

$$t \geq f_\ell(x) \Leftrightarrow \exists u^\ell : A_\ell(x, t, u^\ell) \succeq_{\mathbf{K}_\ell} 0,$$

where \mathbf{K}_ℓ is a direct product of ice-cream cones, and let

$$f : \mathbf{R}^m \rightarrow \mathbf{R} \cup \{+\infty\}$$

be CQr:

$$t \geq f(y) \Leftrightarrow \exists v : A(y, t, v) \succeq_{\mathbf{K}} 0,$$

where \mathbf{K} is a direct product of ice-cream cones.

Assume that f is monotone with respect to the usual partial ordering of \mathbf{R}^m :

$$y' \geq y'' \Rightarrow f(y') \geq f(y''),$$

and that the superposition

$$g(x) = \begin{cases} f(f_1(x), \dots, f_m(x)) & f_\ell(x) < \infty, \ell = 1, \dots, m \\ +\infty & \text{otherwise} \end{cases}$$

is a proper function (i.e., it is finite at least at one point).

Theorem 2.3.1 Under the above setting, the superposition g is CQr with CQR

$$t \geq g(x) \Leftrightarrow \exists t_1, \dots, t_m, u^1, \dots, u^m, v : \begin{cases} A_\ell(x, t_\ell, u^\ell) \succeq_{\mathbf{K}_\ell} 0, \ell = 1, \dots, m \\ A(t_1, \dots, t_m, t, v) \succeq_{\mathbf{K}} 0 \end{cases} \quad (2.3.25)$$

Proof of this simple statement is left to the reader.

Remark 2.3.1 If part of the “inner” functions, say, f_1, \dots, f_k , are affine, it suffices to require the monotonicity of the “outer” function f with respect to the variables y_{k+1}, \dots, y_m only. A CQR for the superposition in this case becomes

$$t \geq g(x) \Leftrightarrow \exists t_{k+1}, \dots, t_m, u^{k+1}, \dots, u^m, v : \begin{cases} A_\ell(x, t_\ell, u^\ell) \succeq_{\mathbf{K}_\ell} 0, \ell = k+1, \dots, m \\ A(f_1(x), f_2(x), \dots, f_k(x), t_{k+1}, t_{k+2}, \dots, t_m, t, v) \succeq_{\mathbf{K}} 0 \end{cases} \quad (2.3.26)$$

2.3.5 More examples of CQ-representable functions/sets

We are sufficiently equipped to build the dictionary of CQ-representable functions/sets. Having built already the “elementary” part of the dictionary, we can add now a more “advanced” part.

7. Convex quadratic form $g(x) = x^T Qx + q^T x + r$ (Q is a positive semidefinite symmetric matrix) is CQr.

Indeed, Q is positive semidefinite symmetric and therefore can be decomposed as $Q = D^T D$, so that $g(x) = \|Dx\|_2^2 + q^T x + r$. We see that g is obtained from our “raw materials” – the squared Euclidean norm and an affine function – by affine substitution of argument and addition.

Here is an explicit CQR of g :

$$\{(x, t) : x^T D^T D x + q^T x + r \leq t\} = \{(x, t) : \left\| \frac{Dx}{\frac{t+q^T x+r}{2}} \right\|_2 \leq \frac{t - q^T x - r}{2}\} \quad (2.3.27)$$

8. The cone $K = \{(x, \sigma_1, \sigma_2) \in \mathbf{R}^n \times \mathbf{R} \times \mathbf{R} : \sigma_1, \sigma_2 \geq 0, \sigma_1 \sigma_2 \geq x^T x\}$ is CQr.

Indeed, the set is just the epigraph of the fractional-quadratic function $x^T x/s$, see Example 5; we simply write σ_1 instead of s and σ_2 instead of t .

Here is an explicit CQR for the set:

$$K = \{(x, \sigma_1, \sigma_2) : \left\| \begin{pmatrix} x \\ \frac{\sigma_1 - \sigma_2}{2} \end{pmatrix} \right\|_2 \leq \frac{\sigma_1 + \sigma_2}{2}\} \quad (2.3.28)$$

Surprisingly, our set is just the ice-cream cone, more precisely, its inverse image under the one-to-one linear mapping

$$\begin{pmatrix} x \\ \sigma_1 \\ \sigma_2 \end{pmatrix} \mapsto \begin{pmatrix} x \\ \frac{\sigma_1 - \sigma_2}{2} \\ \frac{\sigma_1 + \sigma_2}{2} \end{pmatrix}.$$

9. The “half-cone” $K_+^2 = \{(x_1, x_2, t) \in \mathbf{R}^3 : x_1, x_2 \geq 0, 0 \leq t \leq \sqrt{x_1 x_2}\}$ is CQr.

Indeed, our set is the intersection of the cone $\{t^2 \leq x_1 x_2, x_1, x_2 \geq 0\}$ from the previous example and the half-space $t \geq 0$.

Here is the explicit CQR of K_+ :

$$K_+ = \{(x_1, x_2, t) : t \geq 0, \left\| \begin{pmatrix} t \\ \frac{x_1 - x_2}{2} \end{pmatrix} \right\|_2 \leq \frac{x_1 + x_2}{2}\}. \quad (2.3.29)$$

10. The hypograph of the geometric mean – the set $K^2 = \{(x_1, x_2, t) \in \mathbf{R}^3 : x_1, x_2 \geq 0, t \leq \sqrt{x_1 x_2}\}$ – is CQr.

Note the difference with the previous example – here t is not required to be nonnegative!

Here is the explicit CQR for K^2 (cf. Example 9):

$$K^2 = \left\{ (x_1, x_2, t) : \exists \tau : t \leq \tau; \tau \geq 0, \left\| \begin{pmatrix} \tau \\ \frac{x_1 - x_2}{2} \end{pmatrix} \right\|_2 \leq \frac{x_1 + x_2}{2} \right\}.$$

11. The hypograph of the geometric mean of 2^l variables – the set $K^{2^l} = \{(x_1, \dots, x_{2^l}, t) \in \mathbf{R}^{2^l+1} : x_i \geq 0, i = 1, \dots, 2^l, t \leq (x_1 x_2 \dots x_{2^l})^{1/2^l}\}$ – is CQr. To see it and to get its CQR, it suffices to iterate the construction of Example 10. Indeed, let us add to our initial variables a number of additional x -variables:

– let us call our 2^l original x -variables the variables of *level 0* and write $x_{0,i}$ instead of x_i . Let us add one new variable of *level 1* per every two variables of level 0. Thus, we add 2^{l-1} variables $x_{1,i}$ of level 1.

– similarly, let us add one new variable of level 2 per every two variables of level 1, thus adding 2^{l-2} variables $x_{2,i}$; then we add one new variable of level 3 per every two variables of level 2, and so on, until level l with a single variable $x_{l,1}$ is built.

Now let us look at the following system S of constraints:

$$\begin{aligned} \text{layer 1: } & x_{1,i} \leq \sqrt{x_{0,2i-1} x_{0,2i}}, x_{1,i}, x_{0,2i-1}, x_{0,2i} \geq 0, \quad i = 1, \dots, 2^{l-1} \\ \text{layer 2: } & x_{2,i} \leq \sqrt{x_{1,2i-1} x_{1,2i}}, x_{2,i}, x_{1,2i-1}, x_{1,2i} \geq 0, \quad i = 1, \dots, 2^{l-2} \\ & \dots \\ \text{layer } l: & x_{l,1} \leq \sqrt{x_{l-1,1} x_{l-1,2}}, x_{l,1}, x_{l-1,1}, x_{l-1,2} \geq 0 \\ (*) \quad & t \leq x_{l,1} \end{aligned}$$

The inequalities of the first layer say that the variables of the zero and the first level should be nonnegative and every one of the variables of the first level should be \leq the geometric mean of the corresponding pair of our original x -variables. The inequalities of the second layer add the requirement that the variables of the second level should be nonnegative, and every one of them should be \leq the geometric mean of the corresponding pair of the first level variables, etc. It is clear that if all these inequalities and $(*)$ are satisfied, then t is \leq the geometric mean of x_1, \dots, x_{2^l} . Vice versa, given nonnegative x_1, \dots, x_{2^l} and a real t which is \leq the geometric mean of x_1, \dots, x_{2^l} , we always can extend these data to a solution of S . In other words, K^{2^l} is the projection of the solution set of S onto the plane of our original variables x_1, \dots, x_{2^l}, t . It remains to note that the set of solutions of S is CQr (as the intersection of CQr sets $\{(v, p, q, r) \in \mathbf{R}^N \times \mathbf{R}_+^3 : r \leq \sqrt{qp}\}$, see Example 9), so that its projection is also CQr. To get a CQR of K^{2^l} , it suffices to replace the inequalities in S with their conic quadratic equivalents, explicitly given in Example 9.

12. The convex increasing power function $x_+^{p/q}$ of rational degree $p/q \geq 1$ is CQr.

Indeed, given positive integers p, q , $p > q$, let us choose the smallest integer l such that $p \leq 2^l$, and consider the CQr set

$$K^{2^l} = \{(y_1, \dots, y_{2^l}, s) \in \mathbf{R}_+^{2^l+1} : s \leq (y_1 y_2 \dots y_{2^l})^{1/2^l}\}. \quad (2.3.30)$$

Setting $r = 2^l - p$, consider the following affine parameterization of the variables from \mathbf{R}^{2^l+1} by two variables ξ, t :

– s and r first variables y_i are all equal to ξ (note that we still have $2^l - r = p \geq q$ “unused” variables y_i);

– q next variables y_i are all equal to t ;

– the remaining y_i ’s, if any, are all equal to 1.

The inverse image of K^{2^l} under this mapping is CQr and it is the set

$$K = \{(\xi, t) \in \mathbf{R}_+^2 : \xi^{1-r/2^l} \leq t^{q/2^l}\} = \{(\xi, t) \in \mathbf{R}_+^2 : t \geq \xi^{p/q}\}.$$

It remains to note that the epigraph of $x_+^{p/q}$ can be obtained from the CQr set K by operations preserving the CQr property. Specifically, the set $L = \{(x, \xi, t) \in \mathbf{R}^3 : \xi \geq 0, \xi \geq x, t \geq \xi^{p/q}\}$ is the intersection of $\mathbf{K} \times \mathbf{R}$ and the half-space $\{(x, \xi, t) : \xi \geq x\}$ and thus is CQr along with K , and $\text{Epi}\{x_+^{p/q}\}$ is the projection of the CQr set L on the plane of x, t -variables.

13. The decreasing power function $g(x) = \begin{cases} x^{-p/q}, & x > 0 \\ +\infty, & x \leq 0 \end{cases}$ (p, q are positive integers) is CQr.

Same as in Example 12, we choose the smallest integer l such that $2^l \geq p + q$, consider the CQr set (2.3.30) and parameterize affinely the variables y_i, s by two variables (x, t) as follows:

- s and the first $(2^l - p - q)$ y_i 's are all equal to one;
- p of the remaining y_i 's are all equal to x , and the q last of y_i 's are all equal to t .

It is immediately seen that the inverse image of K^{2^l} under the indicated affine mapping is the epigraph of g .

14. The even power function $g(x) = x^{2p}$ on the axis (p positive integer) is CQr.

Indeed, we already know that the sets $P = \{(x, \xi, t) \in \mathbf{R}^3 : x^2 \leq \xi\}$ and $K' = \{(x, \xi, t) \in \mathbf{R}^3 : 0 \leq \xi, \xi^p \leq t\}$ are CQr (both sets are direct products of \mathbf{R} and the sets with already known to us CQR's). It remains to note that the epigraph of g is the projection of $P \cap Q$ onto the (x, t) -plane.

Example 14 along with our combination rules allows to build a CQR for a polynomial $p(x)$ of the form

$$p(x) = \sum_{l=1}^L p_l x^{2l}, \quad x \in \mathbf{R},$$

with nonnegative coefficients.

15. The concave monomial $x_1^{\pi_1} \dots x_n^{\pi_n}$. Let $\pi_1 = \frac{p_1}{p}, \dots, \pi_n = \frac{p_n}{p}$ be positive rational numbers with $\pi_1 + \dots + \pi_n \leq 1$. The function

$$f(x) = -x_1^{\pi_1} \dots x_n^{\pi_n} : \mathbf{R}_+^n \rightarrow \mathbf{R}$$

is CQr.

The construction is similar to the one of Example 12. Let l be such that $2^l \geq p$. We recall that the set

$$Y = \{(y_1, \dots, y_{2^l}, s) : y_1, \dots, y_{2^l}, s : y_1, \dots, y_{2^l} \geq 0, 0 \leq s \leq (y_1 \dots, y_{2^l})^{1/2^l}\}$$

is CQr, and therefore so is its inverse image under the affine mapping

$$(x_1, \dots, x_n, s) \mapsto (\underbrace{x_1, \dots, x_1}_{p_1}, \underbrace{x_2, \dots, x_2}_{p_2}, \dots, \underbrace{x_n, \dots, x_n}_{p_n}, \underbrace{s, \dots, s}_{2^l - p}, \underbrace{1, \dots, 1}_{p - p_1 - \dots - p_n}, s),$$

i.e., the set

$$\begin{aligned} Z &= \{(x_1, \dots, x_n, s) : x_1, \dots, x_n \geq 0, 0 \leq s \leq (x_1^{p_1} \dots x_n^{p_n} s^{2^l - p})^{1/2^l}\} \\ &= \{(x_1, \dots, x_n, s) : x_1, \dots, x_n \geq 0, 0 \leq s \leq x_1^{p_1/p} \dots x_n^{p_n/p}\}. \end{aligned}$$

Since the set Z is CQr, so is the set

$$Z' = \{(x_1, \dots, x_n, t, s) : x_1, \dots, x_n \geq 0, s \geq 0, 0 \leq s - t \leq x_1^{\pi_1} \dots x_n^{\pi_n}\},$$

which is the intersection of the half-space $\{s \geq 0\}$ and the inverse image of Z under the affine mapping $(x_1, \dots, x_n, t, s) \mapsto (x_1, \dots, x_n, s - t)$. It remains to note that the epigraph of f is the projection of Z' onto the plane of the variables x_1, \dots, x_n, t .

16. The convex monomial $x_1^{-\pi_1} \dots x_n^{-\pi_n}$. Let π_1, \dots, π_n be positive rational numbers. The function

$$f(x) = x_1^{-\pi_1} \dots x_n^{-\pi_n} : \{x \in \mathbf{R}^n : x > 0\} \rightarrow \mathbf{R}$$

is CQr.

The verification is completely similar to the one in Example 15.

17. The p -norm $\|x\|_p = \left(\sum_{i=1}^n |x_i|^p \right)^{1/p} : \mathbf{R}^n \rightarrow \mathbf{R}$ ($p \geq 1$ is a rational number). We claim that the function $\|x\|_p$ is CQr.

It is immediately seen that

$$\|x\|_p \leq t \Leftrightarrow t \geq 0 \text{ & } \exists v_1, \dots, v_n \geq 0 : |x_i| \leq t^{(p-1)/p} v_i^{1/p}, i = 1, \dots, n, \sum_{i=1}^n v_i \leq t. \quad (2.3.31)$$

Indeed, if the indicated v_i exist, then $\sum_{i=1}^n |x_i|^p \leq t^{p-1} \sum_{i=1}^n v_i \leq t^p$, i.e., $\|x\|_p \leq t$. Vice versa, assume that $\|x\|_p \leq t$. If $t = 0$, then $x = 0$, and the right hand side relations in (2.3.31) are satisfied for $v_i = 0$, $i = 1, \dots, n$. If $t > 0$, we can satisfy these relations by setting $v_i = |x_i|^p t^{1-p}$.

(2.3.31) says that the epigraph of $\|x\|_p$ is the projection onto the (x, t) -plane of the set of solutions to the system of inequalities

$$\begin{aligned} t &\geq 0 \\ v_i &\geq 0, i = 1, \dots, n \\ x_i &\leq t^{(p-1)/p} v_i^{1/p}, i = 1, \dots, n \\ -x_i &\leq t^{(p-1)/p} v_i^{1/p}, i = 1, \dots, n \\ v_1 + \dots + v_n &\leq t \end{aligned}$$

Each of these inequalities defines a CQr set (in particular, for the nonlinear inequalities this is due to Example 15). Thus, the solution set of the system is CQr (as an intersection of finitely many CQr sets), whence its projection on the (x, t) -plane – i.e., the epigraph of $\|x\|_p$ – is CQr.

17b. The function $\|x_+\|_p = \left(\sum_{i=1}^n \max^p[x_i, 0] \right)^{1/p} : \mathbf{R}^n \rightarrow \mathbf{R}$ ($p \geq 1$ a rational number) is CQr.

Indeed,

$$t \geq \|x_+\|_p \Leftrightarrow \exists y_1, \dots, y_n : 0 \leq y_i, x_i \leq y_i, i = 1, \dots, n, \|y\|_p \leq t.$$

Thus, the epigraph of $\|x_+\|_p$ is a projection of the CQr set (see Example 17) given by the system of inequalities in the right hand side.

From the above examples it is seen that the “expressive abilities” of c.q.i.’s are indeed strong: they allow to handle a wide variety of very different functions and sets.

2.4 More applications: Robust Linear Programming

Equipped with abilities to treat a wide variety of CQr functions and sets, we can consider now an important generic application of Conic Quadratic Programming, specifically, in the *Robust Linear Programming*.

2.4.1 Robust Linear Programming: the paradigm

Consider an LP program

$$\min_x \{c^T x : Ax - b \geq 0\}. \quad (\text{LP})$$

In real world applications, the data c, A, b of (LP) is not always known exactly; what is typically known is a domain \mathcal{U} in the space of data – an “uncertainty set” – which for sure contains the “actual” (unknown) data. There are cases in reality where, in spite of this data uncertainty, our decision x must satisfy the “actual” constraints, whether we know them or not. Assume, e.g., that (LP) is a model of a technological process in Chemical Industry, so that entries of x represent the amounts of different kinds of materials participating in the process. Typically the process includes a number of decomposition-recombination stages. A model of this problem must take care of natural balance restrictions: the amount of every material to be used at a particular stage cannot exceed the amount of the same material yielded by the preceding stages. In a meaningful production plan, these balance inequalities must be satisfied even though they involve coefficients affected by unavoidable uncertainty of the exact contents of the raw materials, of time-varying parameters of the technological devices, etc.

If indeed all we know about the data is that they belong to a given set \mathcal{U} , but we still have to satisfy the actual constraints, the only way to meet the requirements is to restrict ourselves to *robust feasible* candidate solutions – those satisfying *all possible* realizations of the uncertain constraints, i.e., vectors x such that

$$Ax - b \geq 0 \quad \forall [A; b] \text{ such that } \exists c : (c, A, b) \in \mathcal{U}. \quad (2.4.1)$$

In order to choose among these robust feasible solutions the best possible, we should decide how to “aggregate” the various realizations of the objective into a single “quality characteristic”. To be methodologically consistent, we use the same worst-case-oriented approach and take as an objective function $f(x)$ the maximum, over all possible realizations of the objective $c^T x$:

$$f(x) = \sup\{c^T x \mid c : \exists [A; b] : (c, A, b) \in \mathcal{U}\}.$$

With this methodology, we can associate with our *uncertain LP program* (i.e., with the family

$$\mathcal{LP}(\mathcal{U}) = \left\{ \min_{x: Ax \geq b} c^T x \mid (c, A, b) \in \mathcal{U} \right\}$$

of all usual (“certain”) LP programs with the data belonging to \mathcal{U}) its robust counterpart. In the latter problem we are seeking for a robust feasible solution with the smallest possible value of the “guaranteed objective” $f(x)$. In other words, the robust counterpart of $\mathcal{LP}(\mathcal{U})$ is the optimization problem

$$\min_{t, x} \left\{ t : c^T x \leq t, Ax - b \geq 0 \quad \forall (c, A, b) \in \mathcal{U} \right\}. \quad (\text{R})$$

Note that (R) is a usual – “certain” – optimization problem, but typically it is not an LP program: the structure of (R) depends on the geometry of the uncertainty set \mathcal{U} and can be very complicated.

As we shall see in a while, in many cases it is reasonable to specify the uncertainty set \mathcal{U} as an *ellipsoid* – the image of the unit Euclidean ball under an affine mapping – or, more generally, as a CQr set. As we shall see in a while, in this case the robust counterpart of an uncertain LP problem is (equivalent to) an explicit conic quadratic program. Thus, Robust

Linear Programming with CQR uncertainty sets can be viewed as a “generic source” of conic quadratic problems.

Let us look at the robust counterpart of an uncertain LP program

$$\left\{ \min_x \left\{ c^T x : a_i^T x - b_i \geq 0, i = 1, \dots, m \right\} \mid (c, A, b) \in \mathbf{U} \right\}$$

in the case of a “simple” ellipsoidal uncertainty – one where the data (a_i, b_i) of i -th inequality constraint

$$a_i^T x - b_i \geq 0,$$

and the objective c are allowed to run independently of each other through respective ellipsoids E_i, E . Thus, we assume that the uncertainty set is

$$\mathcal{U} = \left\{ (a_1, b_1; \dots; a_m, b_m; c) : \exists (\{u_i, u_i^T u_i \leq 1\}_{i=0}^m) : c = c_* + P_0 u_0, \begin{pmatrix} a_i \\ b_i \end{pmatrix} = \begin{pmatrix} a_i^* \\ b_i^* \end{pmatrix} + P_i u_i, i = 1, \dots, m \right\},$$

where c_*, a_i^*, b_i^* are the “nominal data” and $P_i u_i, i = 0, 1, \dots, m$, represent the data perturbations; the restrictions $u_i^T u_i \leq 1$ enforce these perturbations to vary in ellipsoids.

In order to realize that the robust counterpart of our uncertain LP problem is a conic quadratic program, note that x is robust feasible if and only if for every $i = 1, \dots, m$ we have

$$\begin{aligned} 0 &\leq \min_{u_i: u_i^T u_i \leq 1} \left[a_i^T [u] x - b_i [u] : \begin{pmatrix} a_i[u] \\ b_i[u] \end{pmatrix} = \begin{pmatrix} a_i^* \\ b_i^* \end{pmatrix} + P_i u_i \right] \\ &= (a_i^* x)^T x - b_i^* + \min_{u_i: u_i^T u_i \leq 1} u_i^T P_i^T \begin{pmatrix} x \\ -1 \end{pmatrix} \\ &= (a_i^* x)^T x - b_i^* - \left\| P_i^T \begin{pmatrix} x \\ -1 \end{pmatrix} \right\|_2 \end{aligned}$$

Thus, x is robust feasible if and only if it satisfies the system of c.q.i.’s

$$\left\| P_i^T \begin{pmatrix} x \\ -1 \end{pmatrix} \right\|_2 \leq [a_i^*]^T x - b_i^*, \quad i = 1, \dots, m.$$

Similarly, a pair (x, t) satisfies all realizations of the inequality $c^T x \leq t$ “allowed” by our ellipsoidal uncertainty set \mathcal{U} if and only if

$$c_*^T x + \|P_0^T x\|_2 \leq t.$$

Thus, the robust counterpart (R) becomes the conic quadratic program

$$\min_{x, t} \left\{ t : \|P_0^T x\|_2 \leq -c_*^T x + t; \left\| P_i^T \begin{pmatrix} x \\ -1 \end{pmatrix} \right\|_2 \leq [a_i^*]^T x - b_i^*, \quad i = 1, \dots, m \right\} \quad (\text{RLP})$$

2.4.2 Robust Linear Programming: examples

Example 1: Robust synthesis of antenna array. Consider a monochromatic transmitting antenna placed at the origin. Physics says that

1. The directional distribution of energy sent by the antenna can be described in terms of *antenna’s diagram* which is a complex-valued function $D(\delta)$ of a 3D direction δ . The directional distribution of energy sent by the antenna is proportional to $|D(\delta)|^2$.

2. When the antenna is comprised of several antenna elements with diagrams $D_1(\delta), \dots, D_k(\delta)$, the diagram of the antenna is just the sum of the diagrams of the elements.

In a typical Antenna Design problem, we are given several antenna elements with diagrams $D_1(\delta), \dots, D_k(\delta)$ and are allowed to multiply these diagrams by complex weights x_i (which in reality corresponds to modifying the output powers and shifting the phases of the elements). As a result, we can obtain, as a diagram of the array, any function of the form

$$D(\delta) = \sum_{i=1}^k x_i D_i(\delta),$$

and our goal is to find the weights x_i which result in a diagram as close as possible, in a prescribed sense, to a given “target diagram” $D_*(\delta)$.

Consider an example of a planar antenna comprised of a central circle and 9 concentric rings of the same area as the circle (Fig. 2.1.(a)) in the XY -plane (“Earth’s surface”). Let the wavelength be $\lambda = 50\text{cm}$, and the outer radius of the outer ring be 1 m (twice the wavelength).

One can easily see that the diagram of a ring $\{a \leq r \leq b\}$ in the plane XY (r is the distance from a point to the origin) as a function of a 3-dimensional direction δ depends on the altitude (the angle θ between the direction and the plane) only. The resulting function of θ turns out to be *real-valued*, and its analytic expression is

$$D_{a,b}(\theta) = \frac{1}{2} \int_a^b \left[\int_0^{2\pi} r \cos(2\pi r \lambda^{-1} \cos(\theta) \cos(\phi)) d\phi \right] dr.$$

Fig. 2.1.(b) represents the diagrams of our 10 rings for $\lambda = 50\text{cm}$.

Assume that our goal is to design an array with a real-valued diagram which should be axial symmetric with respect to the Z -axis and should be “concentrated” in the cone $\pi/2 \geq \theta \geq \pi/2 - \pi/12$. In other words, our target diagram is a real-valued function $D_*(\theta)$ of the altitude θ with $D_*(\theta) = 0$ for $0 \leq \theta \leq \pi/2 - \pi/12$ and $D_*(\theta)$ somehow approaching 1 as θ approaches $\pi/2$. The target diagram $D_*(\theta)$ used in this example is given in Fig. 2.1.(c) (the dashed curve).

Finally, let us measure the discrepancy between a synthesized diagram and the target one by the Tschebyshev distance, taken along the equidistant 120-point grid of altitudes, i.e., by the quantity

$$\tau = \max_{\ell=1, \dots, 120} \left| D_*(\theta_\ell) - \sum_{j=1}^{10} x_j \underbrace{D_{r_{j-1}, r_j}(\theta_\ell)}_{D_j(\theta_\ell)} \right|, \quad \theta_\ell = \frac{\ell\pi}{240}.$$

Our design problem is simplified considerably by the fact that the diagrams of our “building blocks” and the target diagram are real-valued; thus, we need no complex numbers, and the problem we should finally solve is

$$\min_{\tau \in \mathbf{R}, x \in \mathbf{R}^{10}} \left\{ \tau : -\tau \leq D_*(\theta_\ell) - \sum_{j=1}^{10} x_j D_j(\theta_\ell) \leq \tau, \ell = 1, \dots, 120 \right\}. \quad (\text{Nom})$$

This is a simple LP program; its optimal solution x^* results in the diagram depicted at Fig. 2.1.(c). The uniform distance between the actual and the target diagrams is ≈ 0.0621 (recall that the target diagram varies from 0 to 1).

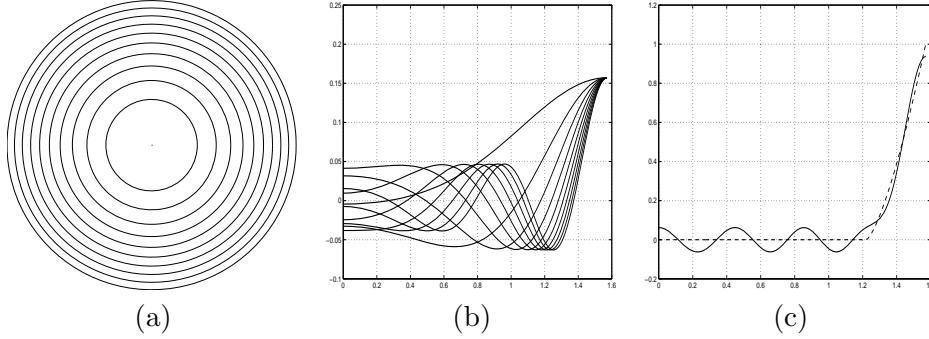


Figure 2.1. Synthesis of antennae array

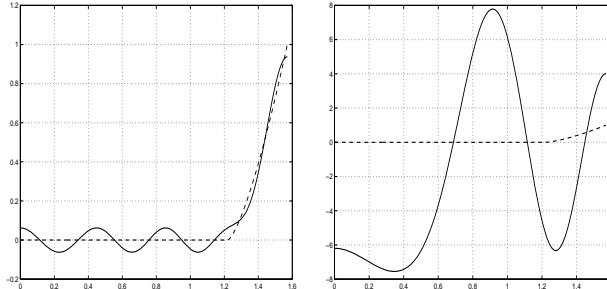
- (a): 10 array elements of equal areas in the XY -plane
the outer radius of the largest ring is 1m, the wavelength is 50cm
- (b): “building blocks” – the diagrams of the rings as functions of the altitude angle θ
- (c): the target diagram (dashed) and the synthesised diagram (solid)

Now recall that our design variables are characteristics of certain physical devices. In reality, of course, we cannot tune the devices to have precisely the optimal characteristics x_j^* ; the best we may hope for is that the actual characteristics x_j^{fct} will coincide with the desired values x_j^* within a small margin, say, 0.1% (this is a fairly high accuracy for a physical device):

$$x_j^{\text{fct}} = p_j x_j^*, \quad 0.999 \leq p_j \leq 1.001.$$

It is natural to assume that the factors p_j are random with the mean value equal to 1; it is perhaps not a great sin to assume that these factors are independent of each other.

Since the actual weights differ from their desired values x_j^* , the actual (random) diagram of our array of antennae will differ from the “nominal” one we see on Fig.2.1.(c). How large could be the difference? Look at the picture:



“Dream and reality”: the nominal (left, solid) and an actual (right, solid) diagrams
[dashed: the target diagram]

The diagram shown to the right is not even the worst case: we just have taken as p_j a sample of 10 independent numbers distributed uniformly in $[0.999, 1.001]$ and have plotted the diagram corresponding to $x_j = p_j x_j^*$. Pay attention not only to the shape (completely opposite to what we need), but also to the scale: the target diagram varies from 0 to 1, and the nominal diagram (the one corresponding to the exact optimal x_j) differs from the target by no more than by 0.0621 (this is the optimal value in the “nominal” problem (Nom)). The actual diagram varies from ≈ -8 to ≈ 8 , and its uniform distance from the target is 7.79 (125 times the nominal optimal value!). We see that our nominal optimal design is completely meaningless: it looks as if we were trying to get the worse possible result, not the best possible one...

How could we get something better? Let us try to apply the Robust Counterpart approach. To this end we take into account from the very beginning that if we want the amplification coefficients to be certain x_j , then the actual amplification coefficients will be $x_j^{\text{fct}} = p_j x_j$, $0.999 \leq p_j \leq 1.001$, and the actual discrepancies will be

$$\delta_\ell(x) = D_*(\theta_\ell) - \sum_{j=1}^{10} p_j x_j D_j(\theta_\ell).$$

Thus, we in fact are solving an uncertain LP problem where the uncertainty affects the coefficients of the constraint matrix (those corresponding to the variables x_j): these coefficients may vary within 0.1% margin of their nominal values.

In order to apply to our uncertain LP program the Robust Counterpart approach, we should specify the uncertainty set \mathcal{U} . The most straightforward way is to say that our uncertainty is “an interval” one – every uncertain coefficient in a given inequality constraint may (independently of all other coefficients) vary through its own uncertainty segment “nominal value $\pm 0.1\%$ ”. This approach, however, is too conservative: we have completely ignored the fact that our p_j ’s are of stochastic nature and are independent of each other, so that it is highly improbable that all of them will simultaneously fluctuate in “dangerous” directions. In order to utilize the statistical independence of perturbations, let us look what happens with a particular inequality

$$-\tau \leq \delta_\ell(x) \equiv D_*(\theta_\ell) - \sum_{j=1}^{10} p_j x_j D_j(\theta_\ell) \leq \tau \quad (2.4.2)$$

when p_j ’s are random. For a fixed x , the quantity $\delta_\ell(x)$ is a random variable with the mean

$$\delta_\ell^*(x) = D_*(\theta_\ell) - \sum_{j=1}^{10} x_j D_j(\theta_\ell)$$

and the standard deviation

$$\begin{aligned} \sigma_\ell(x) &= \sqrt{E\{(\delta_\ell(x) - \delta_\ell^*(x))^2\}} = \sqrt{\sum_{j=1}^{10} x_j^2 D_j^2(\theta_\ell) E\{(p_j - 1)^2\}} \leq \kappa \nu_\ell(x), \\ \nu_\ell(x) &= \sqrt{\sum_{j=1}^{10} x_j^2 D_j^2(\theta_\ell)}, \kappa = 0.001. \end{aligned}$$

Thus, “a typical value” of $\delta_\ell(x)$ differs from $\delta_\ell^*(x)$ by a quantity of order of $\sigma_\ell(x)$. Now let us act as an engineer which believes that a random variable differs from its mean by at most three times its standard deviation; since we are not obliged to be that concrete, let us choose a “safety parameter” ω and ignore all events which result in $|\delta_\ell(x) - \delta_\ell^*(x)| > \omega \nu_\ell(x)$ ⁵⁾. As for the remaining events – those with $|\delta_\ell(x) - \delta_\ell^*(x)| \leq \omega \nu_\ell(x)$ – we take upon ourselves full responsibility. With this approach, a “reliable deterministic version” of the uncertain constraint (2.4.2) becomes the pair of inequalities

$$\begin{aligned} -\tau &\leq \delta_\ell^*(x) - \omega \nu_\ell(x), \\ \delta_\ell^*(x) + \omega \nu_\ell(x) &\leq \tau; \end{aligned}$$

^{5*)} It would be better to use here σ_ℓ instead of ν_ℓ ; however, we did not assume that we know the distribution of p_j , this is why we replace unknown σ_ℓ with its known upper bound ν_ℓ

Replacing all uncertain inequalities in (Nom) with their “reliable deterministic versions” and recalling the definition of $\delta_\ell^*(x)$ and $\nu_\ell(x)$, we end up with the optimization problem

$$\begin{aligned} \text{minimize} & \quad \tau \\ \text{s.t.} & \\ \|Q_\ell x\|_2 & \leq [D_*(\theta_\ell) - \sum_{j=1}^{10} x_j D_j(\theta_\ell)] + \tau, \quad \ell = 1, \dots, 120 \\ \|Q_\ell x\|_2 & \leq -[D_*(\theta_\ell) - \sum_{j=1}^{10} x_j D_j(\theta_\ell)] + \tau, \quad \ell = 1, \dots, 120 \\ & [Q_\ell = \omega \kappa \text{Diag}(D_1(\theta_\ell), D_2(\theta_\ell), \dots, D_{10}(\theta_\ell))] \end{aligned} \quad (\text{Rob})$$

It is immediately seen that (Rob) is nothing but the robust counterpart of (Nom) corresponding to a simple ellipsoidal uncertainty, namely, the one as follows:

The only data of a constraint

$$\sum_{j=1}^{10} A_{\ell j} x_j \leq p_\ell \tau + q_\ell$$

(all constraints in (Nom) are of this form) affected by the uncertainty are the coefficients $A_{\ell j}$ of the left hand side, and the difference $dA[\ell]$ between the vector of these coefficients and the nominal value $(D_1(\theta_\ell), \dots, D_{10}(\theta_\ell))^T$ of the vector of coefficients belongs to the ellipsoid

$$\{dA[\ell] = \omega \kappa Q_\ell u : u \in \mathbf{R}^{10}, u^T u \leq 1\}.$$

Thus, the above “engineering reasoning” leading to (Rob) was nothing but a reasonable way to specify the uncertainty ellipsoids!

The bottom line of our “engineering reasoning” deserves to be formulated as a separate statement and to be equipped with a “reliability bound”:

Proposition 2.4.1 *Consider a randomly perturbed linear constraint*

$$a_0(x) + \epsilon_1 a_1(x) + \dots + \epsilon_n a_n(x) \geq 0, \quad (2.4.3)$$

where $a_j(x)$ are deterministic affine functions of the design vector x , and ϵ_j are independent random perturbations with zero means and such that $|\epsilon_j| \leq \sigma_j$. Assume that x satisfies the “reliable” version of (2.4.3), specifically, the deterministic constraint

$$a_0(x) - \kappa \sqrt{\sigma_1^2 a_1^2(x) + \dots + \sigma_n^2 a_n^2(x)} \geq 0 \quad (2.4.4)$$

($\kappa > 0$). Then x satisfies a realization of (2.4.3) with probability at least $1 - \exp\{-\kappa^2/2\}$.

Proof. All we need is to verify the following Hoeffding’s bound on probabilities of large deviations:

If a_i are deterministic reals and ϵ_i are independent random variables with zero means and such that $|\epsilon_i| \leq \sigma_i$ for given deterministic σ_i , then for every $\kappa \geq 0$ one has

$$p(\kappa) \equiv \text{Prob} \left\{ \sum_i \epsilon_i a_i > \kappa \underbrace{\sqrt{\sum_i a_i^2 \sigma_i^2}}_{\sigma} \right\} \leq \exp\{-\kappa^2/2\}.$$

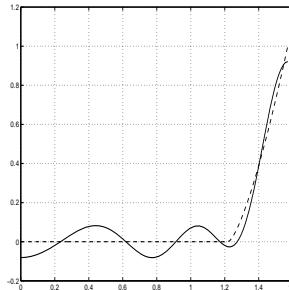
Verification is easy: denoting by \mathbf{E} the expectation, for $\gamma > 0$ we have

$$\begin{aligned} \exp\{\gamma \kappa \sigma\} p(\kappa) &\leq \mathbf{E} \left\{ \exp\left\{ \gamma \sum_i a_i \epsilon_i \right\} \right\} \\ &= \prod_i \mathbf{E} \left\{ \exp\{\gamma a_i \epsilon_i\} \right\} \\ &\quad [\text{since } \epsilon_i \text{ are independent of each other}] \\ &= \prod_i \mathbf{E} \left\{ \exp\{\gamma a_i \epsilon_i\} - \sinh(\gamma a_i \sigma_i) \sigma_i^{-1} \epsilon_i \right\} \\ &\quad [\text{since } \mathbf{E}\{\epsilon_i\} = 0] \\ &\leq \prod_i \max_{-\sigma_i \leq s_i \leq \sigma_i} [\exp\{\gamma a_i s_i\} - \sinh(\gamma a_i \sigma_i) s_i] \\ &= \prod_i \cosh(\gamma a_i \sigma_i) = \prod_i \left[\sum_{k=0}^{\infty} \frac{[\gamma^2 a_i^2 \sigma_i^2]^k}{(2k)!} \right] \\ &\leq \prod_i \left[\sum_{k=0}^{\infty} \frac{[\gamma^2 a_i^2 \sigma_i^2]^k}{2^k k!} \right] \\ &= \prod_i \exp\left\{ \frac{\gamma^2 a_i^2 \sigma_i^2}{2} \right\} \\ &= \exp\{\gamma^2 \sigma^2\}. \end{aligned}$$

Thus,

$$p(\kappa) \leq \min_{\gamma > 0} \exp\left\{ \frac{\gamma^2 \sigma^2}{2} - \gamma \kappa \sigma \right\} = \exp\left\{ -\frac{\kappa^2}{2} \right\}. \quad \blacksquare$$

Now let us look what are the diagrams yielded by the Robust Counterpart approach – i.e., those given by the robust optimal solution. These diagrams are also random (neither the nominal nor the robust solution cannot be implemented exactly!). However, it turns out that they are incomparably closer to the target (and to each other) than the diagrams associated with the optimal solution to the “nominal” problem. Look at a typical “robust” diagram:



A “Robust” diagram. Uniform distance from the target is 0.0822.
[the safety parameter for the uncertainty ellipsoids is $\omega = 1$]

With the safety parameter $\omega = 1$, the robust optimal value is 0.0817; although it is by 30% larger than the nominal optimal value 0.0635, the robust optimal value has a definite advantage that it indeed says something reliable about the quality of actual diagrams we can obtain when implementing the robust optimal solution: in a sample of 40 realizations of the diagrams corresponding to the robust optimal solution, the uniform distances from the target were varying from 0.0814 to 0.0830.

We have built the robust optimal solution under the assumption that the “implementation errors” do not exceed 0.1%. What happens if in reality the errors are larger – say, 1%? It turns out that nothing dramatic happens: now in a sample of 40 diagrams given by the “old” robust optimal solution (affected by 10 times larger “implementation errors”) the uniform distances from the target were varying from 0.0834 to 0.116. Imagine what will happen with the nominal solution under the same circumstances...

The last issue to be addressed here is: why is the nominal solution so unstable? And why with the robust counterpart approach we were able to get a solution which is incomparably better, as far as “actual implementation” is concerned? The answer becomes clear when looking at the nominal and the robust optimal weights:

j	1	2	3	4	5	6	7	8	9	10
x_j^{nom}	1624.4	-14701	55383	-107247	95468	19221	-138622	144870	-69303	13311
x_j^{rob}	-0.3010	4.9638	-3.4252	-5.1488	6.8653	5.5140	5.3119	-7.4584	-8.9140	13.237

It turns out that the nominal problem is “ill-posed” – although its optimal solution is far away from the origin, there is a “massive” set of “nearly optimal” solutions, and among the latter ones we can choose solutions of quite moderate magnitude. Indeed, here are the optimal values obtained when we add to the constraints of (Nom) the box constraints $|x_j| \leq L$, $j = 1, \dots, 10$:

L	1	10	10^2	10^3	10^4	10^5	10^6	10^7
Opt_Val	0.09449	0.07994	0.07358	0.06955	0.06588	0.06272	0.06215	0.06215

Since the “implementation inaccuracies” for a solution are the larger the larger the solution is, there is no surprise that our “huge” nominal solution results in a very unstable actual design. In contrast to this, the Robust Counterpart penalizes the (properly measured) magnitude of x (look at the terms $\|Q_\ell x\|_2$ in the constraints of (Rob)) and therefore yields a much more stable design. Note that this situation is typical for many applications: the nominal solution is on the boundary of the nominal feasible domain, and there are “nearly optimal” solutions to the nominal problem which are in the “deep interior” of this domain. When solving the nominal problem, we do not take any care of a reasonable tradeoff between the “depth of feasibility” and the optimality: any improvement in the objective is sufficient to make the solution just marginally feasible for the nominal problem. And a solution which is only marginally feasible in the nominal problem can easily become “very infeasible” when the data are perturbed. This would not be the case for a “deeply interior” solution. With the Robust Counterpart approach, we do use certain tradeoff between the “depth of feasibility” and the optimality – we are trying to find something like the “deepest feasible nearly optimal solution”; as a result, we normally gain a lot in stability; and if, as in our example, there are “deeply interior nearly optimal” solutions, we do not lose that much in optimality.

Example 2: NETLIB Case Study. NETLIB is a collection of about 100 not very large LPs, mostly of real-world origin, used as the standard benchmark for LP solvers. In the study to

be described, we used this collection in order to understand how “stable” are the feasibility properties of the standard – “nominal” – optimal solutions with respect to small uncertainty in the data. To motivate the methodology of this “Case Study”, here is the constraint # 372 of the problem PILOT4 from NETLIB:

$$\begin{aligned}
 a^T x &\equiv -15.79081x_{826} - 8.598819x_{827} - 1.88789x_{828} - 1.362417x_{829} - 1.526049x_{830} \\
 &\quad - 0.031883x_{849} - 28.725555x_{850} - 10.792065x_{851} - 0.19004x_{852} - 2.757176x_{853} \\
 &\quad - 12.290832x_{854} + 717.562256x_{855} - 0.057865x_{856} - 3.785417x_{857} - 78.30661x_{858} \\
 &\quad - 122.163055x_{859} - 6.46609x_{860} - 0.48371x_{861} - 0.615264x_{862} - 1.353783x_{863} \\
 &\quad - 84.644257x_{864} - 122.459045x_{865} - 43.15593x_{866} - 1.712592x_{870} - 0.401597x_{871} \\
 &\quad + x_{880} - 0.946049x_{898} - 0.946049x_{916} \\
 &\geq b \equiv 23.387405
 \end{aligned} \tag{C}$$

The related nonzero coordinates in the optimal solution x^* of the problem, as reported by CPLEX (one of the best commercial LP solvers), are as follows:

$$\begin{aligned}
 x_{826}^* &= 255.6112787181108 & x_{827}^* &= 6240.488912232100 & x_{828}^* &= 3624.613324098961 \\
 x_{829}^* &= 18.20205065283259 & x_{849}^* &= 174397.0389573037 & x_{870}^* &= 14250.00176680900 \\
 x_{871}^* &= 25910.00731692178 & x_{880}^* &= 104958.3199274139
 \end{aligned}$$

The indicated optimal solution makes (C) an equality within machine precision.

Observe that most of the coefficients in (C) are “ugly reals” like -15.79081 or -84.644257. We have all reasons to believe that coefficients of this type characterize certain technological devices/processes, and as such *they could hardly be known to high accuracy*. It is quite natural to assume that the “ugly coefficients” are in fact uncertain – they coincide with the “true” values of the corresponding data within accuracy of 3-4 digits, not more. The only exception is the coefficient 1 of x_{880} – it perhaps reflects the structure of the problem and is therefore exact – “certain”.

Assuming that the uncertain entries of a are, say, 0.1%-accurate approximations of unknown entries of the “true” vector of coefficients \tilde{a} , we looked what would be the effect of this uncertainty on the validity of the “true” constraint $\tilde{a}^T x \geq b$ at x^* . Here is what we have found:

- The minimum (over all vectors of coefficients \tilde{a} compatible with our “0.1%-uncertainty hypothesis”) value of $\tilde{a}^T x^* - b$, is < -104.9 ; in other words, the violation of the constraint can be as large as 450% of the right hand side!
- Treating the above worst-case violation as “too pessimistic” (why should the true values of all uncertain coefficients differ from the values indicated in (C) in the “most dangerous” way?), consider a more realistic measure of violation. Specifically, assume that the true values of the uncertain coefficients in (C) are obtained from the “nominal values” (those shown in (C)) by random perturbations $a_j \mapsto \tilde{a}_j = (1 + \xi_j)a_j$ with independent and, say, uniformly distributed on $[-0.001, 0.001]$ “relative perturbations” ξ_j . What will be a “typical” relative violation

$$V = \frac{\max[b - \tilde{a}^T x^*, 0]}{b} \times 100\%$$

of the “true” (now random) constraint $\tilde{a}^T x \geq b$ at x^* ? The answer is nearly as bad as for the worst scenario:

Prob{ $V > 0$ }	Prob{ $V > 150\%$ }	Mean(V)
0.50	0.18	125%

Table 2.1. Relative violation of constraint # 372 in PILOT4
(1,000-element sample of 0.1% perturbations of the uncertain data)

We see that quite small (just 0.1%) perturbations of “obviously uncertain” data coefficients can make the “nominal” optimal solution x^* heavily infeasible and thus – practically meaningless.

Inspired by this preliminary experiment, we have carried out the “diagnosis” and the “treatment” phases as follows.

“Diagnosis”. Given a “perturbation level” ϵ (for which we have used the values 1%, 0.1%, 0.01%), for every one of the NETLIB problems, we have measured its “stability index” at this perturbation level, specifically, as follows.

1. We computed the optimal solution x^* of the program by CPLEX.
2. For every one of the inequality constraints

$$a^T x \leq b$$

of the program,

- We looked at the right hand side coefficients a_j and split them into “certain” – those which can be represented, within machine accuracy, as rational fractions p/q with $|q| \leq 100$, and “uncertain” – all the rest. Let J be the set of all uncertain coefficients of the constraint under consideration.
- We defined the *reliability index* of the constraint as the quantity

$$\frac{a^T x^* + \epsilon \sqrt{\sum_{j \in J} a_j^2 (x_j^*)^2} - b}{\max[1, |b|]} \times 100\% \quad (\text{I})$$

Note that the quantity $\epsilon \sqrt{\sum_{j \in J} a_j^2 (x_j^*)^2}$, as we remember from the Antenna story, is of order of typical difference between $a^T x^*$ and $\tilde{a}^T x^*$, where \tilde{a} is obtained from a by random perturbation $a_j \mapsto \tilde{a}_j = p_j a_j$ of uncertain coefficients, with independent random p_j uniformly distributed in the segment $[-\epsilon, \epsilon]$. In other words, *the reliability index is of order of typical violation* (measured in percents of the right hand side) *of the constraint, as evaluated at x^* , under independent random perturbations of uncertain coefficients, ϵ being the relative magnitude of the perturbations.*

3. We treat the nominal solution as *unreliable*, and the problem - as *bad*, the level of perturbations being ϵ , if the worst, over the inequality constraints, reliability index of the constraint is worse than 5%.

The results of the Diagnosis phase of our Case Study were as follows. From the total of 90 NETLIB problems we have processed,

- in 27 problems the nominal solution turned out to be unreliable at the largest ($\epsilon = 1\%$) level of uncertainty;
- 19 of these 27 problems are already bad at the 0.01%-level of uncertainty, and in 13 of these 19 problems, 0.01% perturbations of the uncertain data can make the nominal solution more than 50%-infeasible for some of the constraints.

The details are given in Table 2.2.

Problem	Size ^{a)}	$\epsilon = 0.01\%$		$\epsilon = 0.1\%$		$\epsilon = 1\%$	
		Nbad ^{b)}	Index ^{c)}	Nbad	Index	Nbad	Index
80BAU3B	2263×9799	37	84	177	842	364	8,420
25FV47	822×1571	14	16	28	162	35	1,620
ADLITTLE	57×97			2	6	7	58
AFIRO	28×32			1	5	2	50
BNL2	2325×3489					24	34
BRANDY	221×249					1	5
CAPRI	272×353			10	39	14	390
CYCLE	1904×2857	2	110	5	1,100	6	11,000
D2Q06C	2172×5167	107	1,150	134	11,500	168	115,000
E226	224×282					2	15
FFFFF800	525×854					6	8
FINNIS	498×614	12	10	63	104	97	1,040
GREENBEA	2393×5405	13	116	30	1,160	37	11,600
KB2	44×41	5	27	6	268	10	2,680
MAROS	847×1443	3	6	38	57	73	566
NESM	751×2923					37	20
PEROLD	626×1376	6	34	26	339	58	3,390
PILOT	1442×3652	16	50	185	498	379	4,980
PILOT4	411×1000	42	210,000	63	2,100,000	75	21,000,000
PILOT87	2031×4883	86	130	433	1,300	990	13,000
PILOTJA	941×1988	4	46	20	463	59	4,630
PILOTNOV	976×2172	4	69	13	694	47	6,940
PILOTWE	723×2789	61	12,200	69	122,000	69	1,220,000
SCFXM1	331×457	1	95	3	946	11	9,460
SCFXM2	661×914	2	95	6	946	21	9,460
SCFXM3	991×1371	3	95	9	946	32	9,460
SHARE1B	118×225	1	257	1	2,570	1	25,700

Table 2.2. Bad NETLIB problems.

^{a)} # of linear constraints (excluding the box ones) plus 1 and # of variables

^{b)} # of constraints with index > 5%

^{c)} The worst, over the constraints, reliability index, in %

Our diagnosis leads to the following conclusion:

◊ In real-world applications of Linear Programming one cannot ignore the possibility that a small uncertainty in the data (intrinsic for most real-world LP programs) can make the usual optimal solution of the problem completely meaningless from a practical viewpoint.

Consequently,

◊ In applications of LP, there exists a real need of a technique capable of detecting cases when data uncertainty can heavily affect the quality of the nominal solution, and in these cases to generate a “reliable” solution, one which is immune against uncertainty.

“Treatment”. At the treatment phase of our Case Study, we used the Robust Counterpart methodology, as outlined in Example 1, to pass from “unreliable” nominal solutions of bad NETLIB problems to “uncertainty-immunized” robust solutions. The primary goals here were to understand whether “treatment” is at all possible (the Robust Counterpart may happen to be infeasible) and how “costly” it is – by which margin the robust solution is worse, in terms of

the objective, than the nominal solution. The answers to both these questions turned out to be quite encouraging:

- Reliable solutions do exist, except for the four cases corresponding to the highest ($\epsilon = 1\%$) uncertainty level (see the right column in Table 2.3).
- The price of immunization in terms of the objective value is surprisingly low: when $\epsilon \leq 0.1\%$, it never exceeds 1% and it is less than 0.1% in 13 of 23 cases. Thus, passing to the robust solutions, we gain a lot in the ability of the solution to withstand data uncertainty, while losing nearly nothing in optimality.

The details are given in Table 2.3.

Problem	Nominal optimal value	Objective at robust solution		
		$\epsilon = 0.01\%$	$\epsilon = 0.1\%$	$\epsilon = 1\%$
80BAU3B	987224.2	987311.8 (+ 0.01%)	989084.7 (+ 0.19%)	1009229 (+ 2.23%)
25FV47	5501.846	5501.862 (+ 0.00%)	5502.191 (+ 0.01%)	5505.653 (+ 0.07%)
ADLITTLE	225495.0		225594.2 (+ 0.04%)	228061.3 (+ 1.14%)
AFIRO	-464.7531		-464.7500 (+ 0.00%)	-464.2613 (+ 0.11%)
BNL2	1811.237		1811.237 (+ 0.00%)	1811.338 (+ 0.01%)
BRANDY	1518.511			1518.581 (+ 0.00%)
CAPRI	1912.621		1912.738 (+ 0.01%)	1913.958 (+ 0.07%)
CYCLE	1913.958	1913.958 (+ 0.00%)	1913.958 (+ 0.00%)	1913.958 (+ 0.00%)
D2Q06C	122784.2	122793.1 (+ 0.01%)	122893.8 (+ 0.09%)	Infeasible
E226	-18.75193			-18.75173 (+ 0.00%)
FFFFF800	555679.6			555715.2 (+ 0.01%)
FINNIS	172791.1	172808.8 (+ 0.01%)	173269.4 (+ 0.28%)	178448.7 (+ 3.27%)
GREENBEA	-72555250	-72526140 (+ 0.04%)	-72192920 (+ 0.50%)	-68869430 (+ 5.08%)
KB2	-1749.900	-1749.877 (+ 0.00%)	-1749.638 (+ 0.01%)	-1746.613 (+ 0.19%)
MAROS	-58063.74	-58063.45 (+ 0.00%)	-58011.14 (+ 0.09%)	-57312.23 (+ 1.29%)
NESM	14076040			14172030 (+ 0.68%)
PEROLD	-9380.755	-9380.755 (+ 0.00%)	-9362.653 (+ 0.19%)	Infeasible
PILOT	-557.4875	-557.4538 (+ 0.01%)	-555.3021 (+ 0.39%)	Infeasible
PILOT4	-64195.51	-64149.13 (+ 0.07%)	-63584.16 (+ 0.95%)	-58113.67 (+ 9.47%)
PILOT87	301.7109	301.7188 (+ 0.00%)	302.2191 (+ 0.17%)	Infeasible
PILOTJA	-6113.136	-6113.059 (+ 0.00%)	-6104.153 (+ 0.15%)	-5943.937 (+ 2.77%)
PILOTNOV	-4497.276	-4496.421 (+ 0.02%)	-4488.072 (+ 0.20%)	-4405.665 (+ 2.04%)
PILOTWE	-2720108	-2719502 (+ 0.02%)	-2713356 (+ 0.25%)	-2651786 (+ 2.51%)
SCFXM1	18416.76	18417.09 (+ 0.00%)	18420.66 (+ 0.02%)	18470.51 (+ 0.29%)
SCFXM2	36660.26	36660.82 (+ 0.00%)	36666.86 (+ 0.02%)	36764.43 (+ 0.28%)
SCFXM3	54901.25	54902.03 (+ 0.00%)	54910.49 (+ 0.02%)	55055.51 (+ 0.28%)
SHARE1B	-76589.32	-76589.32 (+ 0.00%)	-76589.32 (+ 0.00%)	-76589.29 (+ 0.00%)

Table 2.3. Objective values for nominal and robust solutions to bad NETLIB problems

2.4.3 Robust counterpart of uncertain LP with a CQr uncertainty set

We have seen that the robust counterpart of uncertain LP with simple “constraint-wise” ellipsoidal uncertainty is a conic quadratic problem. This fact is a special case of the following

Proposition 2.4.2 Consider an uncertain LP

$$\mathcal{LP}(\mathcal{U}) = \left\{ \min_{x:Ax \geq b} c^T x : (c, A, b) \in \mathcal{U} \right\}$$

and assume that the uncertainty set \mathcal{U} is CQR:

$$\mathcal{U} = \{ \zeta = (c, A, B) \in \mathbf{R}^n \times \mathbf{R}^{m \times n} \times \mathbf{R}^m | \exists u : \mathcal{A}(\zeta, u) \equiv P\zeta + Qu + r \geq_{\mathbf{K}} 0 \},$$

where $\mathcal{A}(\zeta, u)$ is an affine mapping and \mathbf{K} is a direct product of ice-cream cones. Assume, further, that the above CQR of \mathcal{U} is strictly feasible:

$$\exists(\bar{\zeta}, \bar{u}) : \mathcal{A}(\bar{\zeta}, \bar{u}) >_{\mathbf{K}} 0.$$

Then the robust counterpart of $\mathcal{LP}(\mathcal{U})$ is equivalent to an explicit conic quadratic problem.

Proof. Introducing an additional variable t and denoting by $z = (t, x)$ the extended vector of design variables, we can write down the instances of our uncertain LP in the form

$$\min_z \left\{ d^T z : \alpha_i^T(\zeta)z - \beta_i(\zeta) \geq 0, i = 1, \dots, m+1 \right\} \quad (\text{LP}[\zeta])$$

with an appropriate vector d ; here the functions

$$\alpha_i(\zeta) = A_i\zeta + a_i, \quad \beta_i(\zeta) = b_i^T \zeta + c_i$$

are affine in the data vector ζ . The robust counterpart of our uncertain LP is the optimization program

$$\min_z \left\{ d^T z \rightarrow \min : \alpha_i^T(\zeta)z - \beta_i(\zeta) \geq 0 \quad \forall \zeta \in \mathcal{U} \quad \forall i = 1, \dots, m+1 \right\}. \quad (\text{RC}_{\text{ini}})$$

Let us fix i and ask ourselves what does it mean that a vector z satisfies the infinite system of linear inequalities

$$\alpha_i^T(\zeta)z - \beta_i(\zeta) \geq 0 \quad \forall \zeta \in \mathcal{U}. \quad (\text{C}_i)$$

Clearly, a given vector z possesses this property if and only if the optimal value in the optimization program

$$\min_{\tau, \zeta} \left\{ \tau : \tau \geq \alpha_i^T(\zeta)z - \beta_i(\zeta), \zeta \in \mathcal{U} \right\}$$

is nonnegative. Recalling the definition of \mathcal{U} , we see that the latter problem is equivalent to the conic quadratic program

$$\min_{\tau, \zeta} \left\{ \tau : \tau \geq \alpha_i^T(\zeta)z - \beta_i(\zeta) \equiv [\underbrace{A_i\zeta + a_i}_{\alpha_i(\zeta)}]^T z - [\underbrace{b_i^T \zeta + c_i}_{\beta_i(\zeta)}], \mathcal{A}(\zeta, u) \equiv P\zeta + Qu + r \geq_{\mathbf{K}} 0 \right\} \quad (\text{CQ}_i[z])$$

in variables τ, ζ, u . Thus, z satisfies (C_i) if and only if the optimal value in $(\text{CQ}_i[z])$ is nonnegative.

Since by assumption the system of conic quadratic inequalities $\mathcal{A}(\zeta, u) \geq_{\mathbf{K}} 0$ is strictly feasible, the conic quadratic program $(\text{CQ}_i[z])$ is strictly feasible. By the Conic Duality Theorem, if (a) the optimal value in $(\text{CQ}_i[z])$ is nonnegative, then (b) the dual to $(\text{CQ}_i[z])$ problem admits a feasible solution with a nonnegative value of the dual objective. By Weak Duality, (b) implies

(a). Thus, the fact that the optimal value in $(CQ_i[z])$ is nonnegative is equivalent to the fact that the dual problem admits a feasible solution with a nonnegative value of the dual objective:

$$\begin{aligned} z \text{ satisfies } (C_i) &\Updownarrow \\ \text{Opt}(CQ_i[z]) \geq 0 &\Updownarrow \\ \left\{ \begin{array}{l} \exists \lambda \in \mathbf{R}, \xi \in \mathbf{R}^N (N \text{ is the dimension of } \mathbf{K}): \\ \lambda[a_i^T z - c_i] - \xi^T r \geq 0, \\ \lambda = 1, \\ -\lambda A_i^T z + b_i + P^T \xi = 0, \\ Q^T \xi = 0, \\ \lambda \geq 0, \\ \xi \geq_{\mathbf{K}} 0. \end{array} \right. &\Updownarrow \\ \left\{ \begin{array}{l} \exists \xi \in \mathbf{R}^N : \\ a_i^T z - c_i - \xi^T r \geq 0 \\ -A_i^T z + b_i + P^T \xi = 0, \\ Q^T \xi = 0, \\ \xi \geq_{\mathbf{K}} 0. \end{array} \right. & \end{aligned}$$

We see that the set of vectors z satisfying (C_i) is CQr :

$$\begin{aligned} z \text{ satisfies } (C_i) &\Updownarrow \\ \left\{ \begin{array}{l} \exists \xi \in \mathbf{R}^N : \\ a_i^T z - c_i - \xi^T r \geq 0 \\ -A_i^T z + b_i + P^T \xi = 0, \\ Q^T \xi = 0, \\ \xi \geq_{\mathbf{K}} 0. \end{array} \right. & \end{aligned}$$

Consequently, the set of robust feasible z – those satisfying (C_i) for all $i = 1, \dots, m+1$ – is CQr (as the intersection of finitely many CQr sets), whence the robust counterpart of our uncertain LP, being the problem of minimizing a linear objective over a CQr set, is equivalent to a conic quadratic problem. Here is this problem:

$$\begin{aligned} & \text{minimize } d^T z \\ \left\{ \begin{array}{l} a_i^T z - c_i - \xi_i^T r \geq 0, \\ -A_i^T z + b_i + P^T \xi_i = 0, \\ Q^T \xi_i = 0, \\ \xi_i \geq_{\mathbf{K}} 0 \end{array} \right. , i = 1, \dots, m+1 & \end{aligned}$$

with design variables $z, \xi_1, \dots, \xi_{m+1}$. Here A_i, a_i, c_i, b_i come from the affine functions $\alpha_i(\zeta) = A_i \zeta + a_i$ and $\beta_i(\zeta) = b_i^T \zeta + c_i$, while P, Q, r come from the description of \mathcal{U} :

$$\mathcal{U} = \{\zeta : \exists u : P\zeta + Qu + r \geq_{\mathbf{K}} 0\}. \quad \blacksquare$$

Remark 2.4.1 Looking at the proof of Proposition 2.4.2, we see that the assumption that the uncertainty set \mathcal{U} is CQr plays no crucial role. What indeed is important is that \mathcal{U} is the projection on the ζ -space of the solution set of a strictly feasible conic inequality associated with certain cone \mathbf{K} . Whenever this is the case, the above construction demonstrates that the robust counterpart of $\mathcal{LP}(\mathcal{U})$ is a conic problem associated with the cone which is a direct product of several cones dual to \mathbf{K} . E.g., when the uncertainty set is polyhedral (i.e., it is given by finitely many scalar linear inequalities: $\mathbf{K} = \mathbf{R}_+^m$), the robust counterpart of $\mathcal{LP}(\mathcal{U})$ is an explicit LP program (and in this case we can eliminate the assumption that the conic inequality defining \mathcal{U} is strictly feasible (why?)). Consider, e.g., an uncertain LP with *interval* uncertainty in the data:

$$\left\{ \min_x \left\{ c^T x : Ax \geq b \right\} : \begin{array}{l} |c_j - c_j^*| \leq \epsilon_j, j = 1, \dots, n \\ A_{ij} \in [A_{ij}^* - \epsilon_{ij}, A_{ij}^* + \epsilon_{ij}], i = 1, \dots, m, j = 1, \dots, n \\ |b_i - b_i^*| \leq \delta_i, i = 1, \dots, m \end{array} \right\}.$$

The (LP equivalent of the) Robust Counterpart of the program is

$$\min_{x,y} \left\{ \sum_j [c_j^* x_j + \epsilon_j y_j] : \begin{array}{l} \sum_j A_{ij}^* x_j - \sum_j \epsilon_{ij} y_j \geq b_i^* + \delta_i, i = 1, \dots, m \\ -y_j \leq x_j \leq y_j, j = 1, \dots, n \end{array} \right\}$$

(why ?)

2.4.4 CQ-representability of the optimal value in a CQ program as a function of the data

Let us ask ourselves the following question: consider a conic quadratic program

$$\min_x \left\{ c^T x : Ax - b \geq_{\mathbf{K}} 0 \right\}, \quad (2.4.5)$$

where \mathbf{K} is a direct product of ice-cream cones and A is a matrix with trivial null space. The optimal value of the problem clearly is a function of the data (c, A, b) of the problem. What can be said about CQ-representability of this function? In general, not much: the function is not even convex. There are, however, two modifications of our question which admit good answers. Namely, under mild regularity assumptions

(a) With c, A fixed, the optimal value is a CQ-representable function of the right hand side vector b ;

(b) with A, b fixed, the minus optimal value is a CQ-representable function of c .

Here are the exact forms of our claims:

Proposition 2.4.3 *Let c, A be fixed, and let \mathcal{B} be a CQr set in $\mathbf{R}^{\dim b}$ such that for every $b \in \mathcal{B}$ problem (2.4.5) is strictly feasible. Then the optimal value of the problem is a CQr function on \mathcal{B} .*

The statement is quite evident: if b is such that (2.4.5) is strictly feasible, then the optimal value $\text{Opt}(b)$ in the problem is either $-\infty$, or is achieved (by Conic Duality Theorem). In both cases,

$$\text{Opt}(b) \leq t \Leftrightarrow \exists x : \left\{ \begin{array}{l} c^T x \leq t, \\ Ax - b \geq_{\mathbf{K}} 0 \end{array} \right.,$$

which is, essentially, a CQR for certain function which coincides with $\text{Opt}(b)$ on the set \mathcal{B} of values of b ; in this CQR, b, t are the “variables of interest”, and x plays the role of the additional variable. The CQR of the function $\text{Opt}(b)$ with the domain \mathcal{B} is readily given, via calculus of CQR’s, by the representation

$$\{b \in \mathcal{B} \& \text{Opt}(b) \leq t\} \Leftrightarrow \exists x : \begin{cases} c^T x \leq t, \\ Ax - b \geq_{\mathbf{K}} 0, \\ b \in \mathcal{B} \end{cases}$$

(recall that \mathcal{B} was assumed to be CQr). ■

The claim (b) is essentially a corollary of (a) – via duality, the optimal value in (2.4.5) is, up to pathological cases, the same as the optimal value in the dual problem, in which c becomes the right hand side vector. Here is the exact formulation:

Proposition 2.4.4 *Let A, b be such that (2.4.5) is strictly feasible. Then the minus optimal value $-\text{Opt}(c)$ of the problem is a CQr function of c .*

Proof. For every c and t , the relation $-\text{Opt}(c) \leq t$ says exactly that (2.4.5) is below bounded with the optimal value $\geq -t$. By the Conic Duality Theorem (note that (2.4.5) is strictly feasible!) this is the case if and only if the dual problem admits a feasible solution with the value of the dual objective $\geq -t$, so that

$$\text{Opt}(c) \leq t \Leftrightarrow \exists y : \begin{cases} b^T y \geq t, \\ A^T y = c, \\ y \geq_{\mathbf{K}} 0. \end{cases}$$

The resulting description of the epigraph of the function $-\text{Opt}(c)$ is a CQR for this function, with c, t playing the role of the “variables of interest” and y being the additional variable. ■
A careful reader could have realized that Proposition 2.4.2 is nothing but a straightforward application of Proposition 2.4.4.

Remark 2.4.2 Same as in the case of Proposition 2.4.2, in Propositions 2.4.3, 2.4.4 the assumption that uncertainty set \mathcal{U} is CQr plays no crucial role. Thus, *Propositions 2.4.3, 2.4.4 remain valid for an arbitrary conic program, up to the fact that in this general case we should speak about the representability of the epigraphs of $\text{Opt}(b)$ and $-\text{Opt}(c)$ via conic inequalities associated with direct products of cones \mathbf{K} , and their duals, rather than about CQ-representability. In particular, Propositions 2.4.2, 2.4.3, 2.4.4 remain valid in the case of Semidefinite representability to be discussed in Lecture 3.*

2.4.5 Affinely Adjustable Robust Counterpart

The rationale behind our Robust Optimization paradigm is based on the following tacit assumptions:

1. All constraints are “a must”, so that a meaningful solution should satisfy all realizations of the constraints from the uncertainty set.
2. All decisions are made in advance and thus cannot tune themselves to the “true” values of the data. Thus, candidate solutions must be fixed vectors, and not functions of the true data.

Here we preserve the first of these two assumptions and try to relax the second of them. The motivation is twofold:

- There are situations in dynamical decision-making when the decisions should be made at subsequent time instants, and decision made at instant t in principle can depend on the part of uncertain data which becomes known at this instant.
- There are situations in LP when some of the decision variables do not correspond to actual decisions; they are artificial “analysis variables” added to the problem in order to convert it to a desired form, say, a Linear Programming one. The analysis variables clearly may adjust themselves to the true values of the data.

To give an example, consider the problem where we look for the best, in the discrete L_1 -norm, approximation of a given sequence b by a linear combination of given sequences a_j , $j = 1, \dots, n$, so that the problem with no data uncertainty is

$$\begin{aligned} & \min_{x,t} \left\{ t : \sum_{t=1}^T |b_t - \sum_j a_{tj} x_j| \leq t \right\} \\ & \Downarrow \\ & \min_{t,x,y} \left\{ t : \sum_{t=1}^T y_t \leq t, -y_t \leq b_t - \sum_j a_{tj} x_j \leq y_t, 1 \leq t \leq T \right\} \quad (\text{LP}) \end{aligned} \quad (\text{P})$$

Note that (LP) is an equivalent LP reformulation of (P), and y are typical analysis variables; whether x 's do or do not represent “actual decisions”, y 's definitely do not represent them. Now assume that the data become uncertain. Perhaps we have reasons to require from (t, x) 's to be independent of actual data and to satisfy the constraint in (P) for all realizations of the data. This requirement means that the variables t, x in (LP) must be data-independent, but we have absolutely no reason to insist on data-independence of y 's: (t, x) is robust feasible for (P) if and only if (t, x) , for all realizations of the data from the uncertainty set, can be extended, by a properly chosen *and perhaps depending on the data vector y* , to a feasible solution of (the corresponding realization of) (LP). In other words, *equivalence between (P) and (LP) is restricted to the case of certain data only; when the data become uncertain, the robust counterpart of (LP) is more conservative than the one of (P)*.

In order to take into account a possibility for (part of) the variables to adjust themselves to the true values of (part of) the data, we could act as follows.

Adjustable and non-adjustable decision variables. Consider an uncertain LP program. Without loss of generality, we may assume that the data are affinely parameterized by properly chosen “perturbation vector” ζ running through a given perturbation set \mathcal{Z} ; thus, our uncertain LP can be represented as the family of LP instances

$$\mathcal{LP} = \left\{ \min_x \{ c^T[\zeta]x : A[\zeta]x - b[\zeta] \geq 0 \} : \zeta \in \mathcal{Z} \right\}$$

Now assume that decision variable x_j is allowed to depend on part of the true data. Since the true data are affine functions of ζ , this is the same as to assume that x_j can depend on “a part” $P_j \zeta$ of the perturbation vector, where P_j is a given matrix. The case of $P_j = 0$ correspond to “here and now” decisions x_j – those which should be done in advance; we shall call these decision variables *non-adjustable*. The case of nonzero P_j (“adjustable decision variable”) corresponds to allowing certain dependence of x_j on the data, and the case when P_j has trivial kernel means that x_j is allowed to depend on the entire true data.

Adjustable Robust Counterpart of \mathcal{LP} . With our assumptions, a natural modification of the Robust Optimization methodology results in the following *adjustable Robust Counterpart of \mathcal{LP}* :

$$\min_{t, \{\phi_j(\cdot)\}_{j=1}^n} \left\{ t : \begin{array}{l} \sum_{j=1}^n c_j[\zeta] \phi_j(P_j \zeta) \leq t \quad \forall \zeta \in \mathcal{Z} \\ \sum_{j=1}^n \phi_j(P_j \zeta) A_j[\zeta] - b[\zeta] \geq 0 \quad \forall \zeta \in \mathcal{Z} \end{array} \right\} \quad (\text{ARC})$$

Here $c_j[\zeta]$ is j -th entry of the objective vector, and $A_j[\zeta]$ is j -th column of the constraint matrix.

It should be stressed that *the variables in (ARC) corresponding to adjustable decision variables in the original problem are not reals; they are “decision rules” – real-valued functions of the corresponding portion $P_j\zeta$ of the data*. This fact makes (ARC) infinite-dimensional optimization problem and thus problem which is extremely difficult for numerical processing. Indeed, in general it is unclear how to represent in a tractable way a general-type function of three (not speaking of three hundred) variables; and how could we hope to find, in an efficient manner, *optimal* decision rules when we even do not know how to write them down? Thus, in general (ARC) has no actual meaning – basically all we can do with the problem is to write it down on paper and then look at it...

2.4.5.1 Affinely Adjustable Robust Counterpart of \mathcal{LP}

A natural way to overcome the outlined difficulty is to restrict the decision rules to be “easily representable”, specifically, *to be affine functions of the allowed portions of data*:

$$\phi_j(P_j\zeta) = \mu_j + \nu_j^T P_j \zeta.$$

With this approach, *our new decision variables become reals μ_j and vectors ν_j , and (ARC) becomes the following problem (called Affinely Adjustable Robust Counterpart of \mathcal{LP})*:

$$\min_{t, \{\mu_j, \nu_j\}_{j=1}^n} \left\{ t : \begin{array}{l} \sum_j c_j[z][\mu_j + \nu_j^T P_j \zeta] \leq t \quad \forall \zeta \in \mathcal{Z} \\ \sum_j [\mu_j + \nu_j^T P_j] A_j[\zeta] - b[\zeta] \geq 0 \quad \forall \zeta \in \mathcal{Z} \end{array} \right\} \quad (\text{AARC})$$

Note that the AARC is “in-between” the usual non-adjustable RC (no dependence of variables on the true data at all) and the ARC (arbitrary dependencies of the decision variables on the allowed portions of the true data). Note also that the *only* reason to restrict ourselves with affine decision rules is the desire to end up with a “tractable” robust counterpart, and even this natural goal for the time being is not achieved. Indeed, the constraints in (AARC) are affine in our new decision variables t, μ_j, ν_j , which is a good news. At the same time, they are *semi-infinite*, same as in the case of the non-adjustable Robust Counterpart, but, in contrast to this latter case, in general are *quadratic* in perturbations rather than to be linear in them. This indeed makes a difference: as we know from Proposition 2.4.2, the usual – non-adjustable – RC of an uncertain LP with CQr uncertainty set is equivalent to an explicit Conic Quadratic problem and as such is computationally tractable (in fact, the latter remain true for the case of non-adjustable RC of uncertain LP with arbitrary “computationally tractable” uncertainty set). In contrast to this, AARC can become intractable for uncertainty sets as simple as boxes. There are, however, good news on AARCs:

- First, there exist a generic “good case” where the AARC is tractable. This is the “fixed recourse” case, where the coefficients of adjustable variables x_j – those with $P_j \neq 0$ – are certain (not affected by uncertainty). In this case, the left hand sides of the constraints in (AARC) are affine in ζ , and thus AARC, same as the usual non-adjustable RC, is computationally tractable whenever the perturbation set \mathcal{Z} is so; in particular, Proposition 2.4.2 remains valid for both RC and AARC.
- Second, we shall see in Lecture 3 that even when AARC is intractable, it still admits tight, in certain precise sense, tractable approximations.

2.4.5.2 Example: Uncertain Inventory Management Problem

The model. Consider a single product inventory system comprised of a warehouse and I factories. The planning horizon is T periods. At a period t :

- d_t is the demand for the product. All the demand must be satisfied;
- $v(t)$ is the amount of the product in the warehouse at the beginning of the period ($v(1)$ is given);

- $p_i(t)$ is the i -th order of the period – the amount of the product to be produced during the period by factory i and used to satisfy the demand of the period (and, perhaps, to replenish the warehouse);
- $P_i(t)$ is the maximal production capacity of factory i ;
- $c_i(t)$ is the cost of producing a unit of the product at a factory i .

Other parameters of the problem are:

- V_{\min} - the minimal allowed level of inventory at the warehouse;
- V_{\max} - the maximal storage capacity of the warehouse;
- Q_i - the maximal cumulative production capacity of i 'th factory throughout the planning horizon.

The goal is to minimize the total production cost over all factories and the entire planning period. When all the data are certain, the problem can be modelled by the following linear program:

$$\begin{aligned} \min_{p_i(t), v(t), F} \quad & F \\ \text{s.t.} \quad & \sum_{t=1}^T \sum_{i=1}^I c_i(t)p_i(t) \leq F \\ & 0 \leq p_i(t) \leq P_i(t), i = 1, \dots, I, t = 1, \dots, T \\ & \sum_{t=1}^T p_i(t) \leq Q(i), i = 1, \dots, I \\ & v(t+1) = v(t) + \sum_{i=1}^I p_i(t) - d_t, t = 1, \dots, T \\ & V_{\min} \leq v(t) \leq V_{\max}, t = 2, \dots, T+1. \end{aligned} \tag{2.4.6}$$

Eliminating v -variables, we get an inequality constrained problem:

$$\begin{aligned} \min_{p_i(t), F} \quad & F \\ \text{s.t.} \quad & \sum_{t=1}^T \sum_{i=1}^I c_i(t)p_i(t) \leq F \\ & 0 \leq p_i(t) \leq P_i(t), i = 1, \dots, I, t = 1, \dots, T \\ & \sum_{t=1}^T p_i(t) \leq Q(i), i = 1, \dots, I \\ & V_{\min} \leq v(1) + \sum_{s=1}^t \sum_{i=1}^I p_i(s) - \sum_{s=1}^t d_s \leq V_{\max}, t = 1, \dots, T. \end{aligned} \tag{2.4.7}$$

Assume that the decision on supplies $p_i(t)$ is made at the beginning of period t , and that we are allowed to make these decisions on the basis of demands d_r observed at periods $r \in I_t$, where I_t is a given subset of $\{1, \dots, t\}$. Further, assume that we should specify our supply policies before the planning period starts (“at period 0”), and that when specifying these policies, we do not know exactly the future demands; all we know is that

$$d_t \in [d_t^* - \theta d_t^*, d_t^* + \theta d_t^*], t = 1, \dots, T, \tag{2.4.8}$$

with given positive θ and positive nominal demand d_t^* . We have now an uncertain LP, where the uncertain data are the actual demands d_t , the decision variables are the supplies $p_i(t)$, and these decision variables are allowed to depend on the data $\{d_\tau : \tau \in I_t\}$ which become known when $p_i(t)$ should be specified. Note that our uncertain LP is a “fixed recourse” one – the uncertainty affects solely the right hand side. Thus, the AARC of the problem is computationally tractable, which is good. Let us build the AARC. Restricting our decision-making policy with *affine decision rules*

$$p_i(t) = \pi_{i,t}^0 + \sum_{r \in I_t} \pi_{i,t}^r d_r, \tag{2.4.9}$$

where the coefficients $\pi_{i,t}^r$ are our new non-adjustable design variables, we get from (2.4.7) the following uncertain Linear Programming problem in variables $\pi_{i,t}^s, F$:

$$\begin{aligned} \min_{\pi, F} \quad & F \\ \text{s.t.} \quad & \sum_{t=1}^T \sum_{i=1}^I c_i(t) \left(\pi_{i,t}^0 + \sum_{r \in I_t} \pi_{i,t}^r d_r \right) \leq F \\ & 0 \leq \pi_{i,t}^0 + \sum_{r \in I_t} \pi_{i,t}^r d_r \leq P_i(t), \quad i = 1, \dots, I, \quad t = 1, \dots, T \\ & \sum_{t=1}^T \left(\pi_{i,t}^0 + \sum_{r \in I_t} \pi_{i,t}^r d_r \right) \leq Q(i), \quad i = 1, \dots, I \\ & V_{\min} \leq v(1) + \sum_{s=1}^t \left(\sum_{i=1}^I \pi_{i,s}^0 + \sum_{r \in I_s} \pi_{i,s}^r d_r \right) - \sum_{s=1}^t d_s \leq V_{\max}, \\ & \quad t = 1, \dots, T \\ & \forall \{d_t \in [d_t^* - \theta d_t^*, d_t^* + \theta d_t^*], \quad t = 1, \dots, T\}, \end{aligned} \tag{2.4.10}$$

or, which is the same,

$$\begin{aligned} \min_{\pi, F} \quad & F \\ \text{s.t.} \quad & \sum_{t=1}^T \sum_{i=1}^I c_i(t) \pi_{i,t}^0 + \sum_{r=1}^T \left(\sum_{i=1}^I \sum_{t:r \in I_t} c_i(t) \pi_{i,t}^r \right) d_r - F \leq 0 \\ & \pi_{i,t}^0 + \sum_{r \in I_t} \pi_{i,t}^r d_r \leq P_i(t), \quad i = 1, \dots, I, \quad t = 1, \dots, T \\ & \pi_{i,t}^0 + \sum_{r \in I_t} \pi_{i,t}^r d_r \geq 0, \quad i = 1, \dots, I, \quad t = 1, \dots, T \\ & \sum_{t=1}^T \pi_{i,t}^0 + \sum_{r=1}^T \left(\sum_{t:r \in I_t} \pi_{i,t}^r \right) d_r \leq Q_i, \quad i = 1, \dots, I \\ & \sum_{s=1}^t \sum_{i=1}^I \pi_{i,s}^0 + \sum_{r=1}^t \left(\sum_{i=1}^I \sum_{s \leq t, r \in I_s} \pi_{i,s}^r - 1 \right) d_r \leq V_{\max} - v(1) \\ & \quad t = 1, \dots, T \\ & - \sum_{s=1}^t \sum_{i=1}^I \pi_{i,s}^0 - \sum_{r=1}^t \left(\sum_{i=1}^I \sum_{s \leq t, r \in I_s} \pi_{i,s}^r - 1 \right) d_r \leq v(1) - V_{\min} \\ & \quad t = 1, \dots, T \\ & \forall \{d_t \in [d_t^* - \theta d_t^*, d_t^* + \theta d_t^*], \quad t = 1, \dots, T\}. \end{aligned} \tag{2.4.11}$$

Now, using the following equivalences

$$\begin{aligned} \sum_{t=1}^T d_t x_t \leq y, \quad \forall d_t \in [d_t^*(1-\theta), d_t^*(1+\theta)] \\ \Updownarrow \\ \sum_{t:x_t < 0} d_t^*(1-\theta)x_t + \sum_{t:x_t > 0} d_t^*(1+\theta)x_t \leq y \\ \Updownarrow \\ \sum_{t=1}^T d_t^* x_t + \theta \sum_{t=1}^T d_t^* |x_t| \leq y, \end{aligned}$$

and defining additional variables

$$\alpha_r \equiv \sum_{t:r \in I_t} c_i(t) \pi_{i,t}^r; \quad \delta_i^r \equiv \sum_{t:r \in I_t} \pi_{i,t}^r; \quad \xi_t^r \equiv \sum_{i=1}^I \sum_{s \leq t, r \in I_s} \pi_{i,s}^r - 1,$$

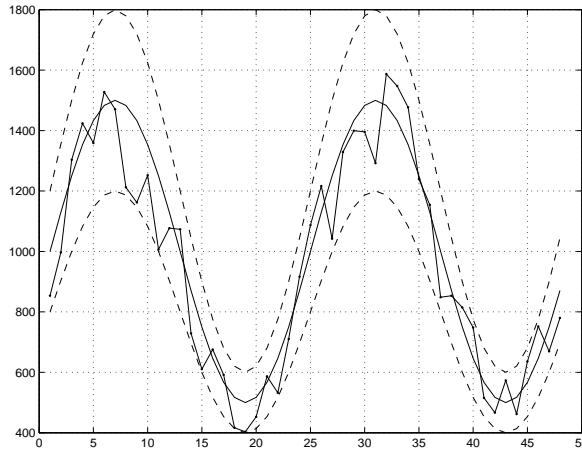
we can straightforwardly convert the AARC (2.4.11) into an equivalent LP (cf. Remark 2.4.1):

$$\begin{aligned}
 & \min_{\pi, F, \alpha, \beta, \gamma, \delta, \zeta, \xi, \eta} F \\
 & \sum_{i=1}^I \sum_{t:r \in I_t} c_i(t) \pi_{i,t}^r = \alpha_r, \quad -\beta_r \leq \alpha_r \leq \beta_r, \quad 1 \leq r \leq T, \quad \sum_{t=1}^T \sum_{i=1}^I c_i(t) \pi_{i,t}^0 + \sum_{r=1}^T \alpha_r d_r^* + \theta \sum_{r=1}^T \beta_r d_r^* \leq F; \\
 & -\gamma_{i,t}^r \leq \pi_{i,t}^r \leq \gamma_{i,t}^r, \quad r \in I_t, \quad \pi_{i,t}^0 + \sum_{r \in I_t} \pi_{i,t}^r d_r^* + \theta \sum_{r \in I_t} \gamma_{i,t}^r d_r^* \leq P_i(t), \quad 1 \leq i \leq I, \quad 1 \leq t \leq T; \\
 & \pi_{i,t}^0 + \sum_{r \in I_t} \pi_{i,t}^r d_r^* - \theta \sum_{r \in I_t} \gamma_{i,t}^r d_r^* \geq 0, \quad \sum_{t:r \in I_t} \pi_{i,t}^r = \delta_i^r, \quad -\zeta_i^r \leq \delta_i^r \leq \zeta_i^r, \quad 1 \leq i \leq I, \quad 1 \leq r \leq T, \\
 & \sum_{t=1}^T \pi_{i,t}^0 + \sum_{r=1}^T \delta_i^r d_r^* + \theta \sum_{r=1}^T \zeta_i^r d_r^* \leq Q_i, \quad 1 \leq i \leq I; \\
 & \sum_{i=1}^I \sum_{s \leq t, r \in I_s} \pi_{i,s}^r - \xi_t^r = 1, \quad -\eta_t^r \leq \xi_t^r \leq \eta_t^r, \quad 1 \leq r \leq t \leq T, \\
 & \sum_{s=1}^t \sum_{i=1}^I \pi_{i,s}^0 + \sum_{r=1}^t \xi_t^r d_r^* + \theta \sum_{r=1}^t \eta_t^r d_r^* \leq V_{\max} - v(1), \quad 1 \leq t \leq T, \\
 & \sum_{s=1}^t \sum_{i=1}^I \pi_{i,s}^0 + \sum_{r=1}^t \xi_t^r d_r^* - \theta \sum_{r=1}^t \eta_t^r d_r^* \geq v(1) - V_{\min}, \quad 1 \leq t \leq T.
 \end{aligned} \tag{2.4.12}$$

An illustrative example. There are $I = 3$ factories producing a seasonal product, and one warehouse. The decisions concerning production are made every two weeks, and we are planning production for 48 weeks, thus the time horizon is $T = 24$ periods. The nominal demand d^* is seasonal, reaching its maximum in winter, specifically,

$$d_t^* = 1000 \left(1 + \frac{1}{2} \sin \left(\frac{\pi(t-1)}{12} \right) \right), \quad t = 1, \dots, 24.$$

We assume that the uncertainty level θ is 20%, i.e., $d_t \in [0.8d_t^*, 1.2d_t^*]$, as shown on the picture.



- Nominal demand (solid)
- “demand tube” – nominal demand $\pm 20\%$ (dashed)
- a sample realization of actual demand (dotted)

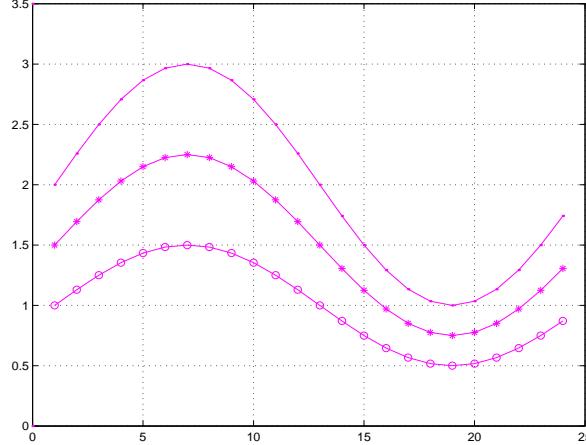
Demand

The production costs per unit of the product depend on the factory and on time and follow the same seasonal pattern as the demand, i.e., rise in winter and fall in summer. The production cost for a factory i at a period t

is given by:

$$c_i(t) = \alpha_i \left(1 + \frac{1}{2} \sin \left(\frac{\pi(t-1)}{12} \right) \right), \quad t = 1, \dots, 24.$$

$$\begin{aligned}\alpha_1 &= 1 \\ \alpha_2 &= 1.5 \\ \alpha_3 &= 2\end{aligned}$$



Production costs for the 3 factories

The maximal production capacity of each one of the factories at each two-weeks period is $P_i(t) = 567$ units, and the integral production capacity of each one of the factories for a year is $Q_i = 13600$. The inventory at the warehouse should not be less than 500 units, and cannot exceed 2000 units.

With this data, the AARC (2.4.12) of the uncertain inventory problem is an LP, the dimensions of which vary, depending on the “information basis” (see below), from 919 variables and 1413 constraints (empty information basis) to 2719 variables and 3213 constraints (on-line information basis).

The experiments. In every one of the experiments, the corresponding management policy was tested against a given number (100) of simulations; in every one of the simulations, the actual demand d_t of period t was drawn at random, according to the uniform distribution on the segment $[(1 - \theta)d_t^*, (1 + \theta)d_t^*]$ where θ was the “uncertainty level” characteristic for the experiment. The demands of distinct periods were independent of each other.

We have conducted two series of experiments:

1. The aim of the first series of experiments was to check the influence of the demand uncertainty θ on the total production costs corresponding to the *robustly adjustable* management policy – the policy (2.4.9) yielded by the optimal solution to the AARC (2.4.12). We compared this cost to the “ideal” one, i.e., the cost we would have paid in the case when all the demands were known to us in advance and we were using the corresponding optimal management policy as given by the optimal solution of (2.4.6).
2. The aim of the second series of experiments was to check the influence of the “information basis” allowed for the management policy, on the resulting management cost. Specifically, in our model as described in the previous Section, when making decisions $p_i(t)$ at time period t , we can make these decisions depending on the demands of periods $r \in I_t$, where I_t is a given subset of the segment $\{1, 2, \dots, t\}$. The larger are these subsets, the more flexible can be our decisions, and hopefully the less are the corresponding management costs. In order to quantify this phenomenon, we considered 4 “information bases” of the decisions:
 - (a) $I_t = \{1, \dots, t\}$ (the richest “on-line” information basis);
 - (b) $I_t = \{1, \dots, t-1\}$ (this *standard information basis* seems to be the most natural “information basis”: past is known, present and future are unknown);
 - (c) $I_t = \{1, \dots, t-4\}$ (the information about the demand is received with a four-day delay);
 - (d) $I_t = \emptyset$ (i.e., no adjusting of future decisions to actual demands at all. This “information basis” corresponds exactly to the management policy yielded by the usual RC of our uncertain LP.).

The results of our experiments are as follows:

1. The influence of the uncertainty level on the management cost. Here we tested the robustly adjustable management policy with the standard information basis against different levels of uncertainty, specifically, the levels of 20%, 10%, 5% and 2.5%. For every uncertainty level, we have computed the average (over 100 simulations) management costs when using the corresponding robustly adaptive management policy. We saved the simulated demand trajectories and then used these trajectories to compute the ideal management costs. The results are summarized in the table below. As expected, the less is the uncertainty, the closer are our management costs to the ideal ones. What is surprising, is the low “price of robustness”: even at the 20% uncertainty level, the average management cost for the robustly adjustable policy was just by 3.4% worse than the corresponding ideal cost; the similar quantity for 2.5%-uncertainty in the demand was just 0.3%.

Uncertainty	AARC		Ideal case		price of robustness
	Mean	Std	Mean	Std	
2.5%	33974	190	33878	194	0.3%
5%	34063	432	33864	454	0.6%
10%	34471	595	34009	621	1.6%
20%	35121	1458	33958	1541	3.4%

Management costs vs. uncertainty level

2. The influence of the information basis. The influence of the information basis on the performance of the robustly adjustable management policy is displayed in the following table:

information basis for decision $p_i(t)$ is demand in periods	Management cost	
	Mean	Std
1, ..., t	34583	1475
1, ..., $t - 1$	35121	1458
1, ..., $t - 4$	Infeasible	
\emptyset	Infeasible	

These experiments were carried out at the uncertainty level of 20%. We see that the poorer is the information basis of our management policy, the worse are the results yielded by this policy. In particular, with 20% level of uncertainty, there does not exist a robust *non-adjustable* management policy: the usual RC of our uncertain LP is infeasible. In other words, in our illustrating example, passing from a priori decisions yielded by RC to “adjustable” decisions yielded by AARC is indeed crucial.

An interesting question is what is the uncertainty level which still allows for a priori decisions. It turns out that the RC is infeasible even at the 5% uncertainty level. Only at the uncertainty level as small as 2.5% the RC becomes feasible and yields the following management costs:

Uncertainty	RC		Ideal cost		price of robustness
	Mean	Std	Mean	Std	
2.5%	35287	0	33842	172	4.3%

Note that even at this unrealistically small uncertainty level the price of robustness for the policy yielded by the RC is by 4.3% larger than the ideal cost (while for the robustly adjustable management this difference is just 0.3%).

Comparison with Dynamic Programming. An Inventory problem we have considered is a typical example of sequential decision-making under dynamical uncertainty, where the information basis for the decision x_t made at time t is the part of the uncertainty revealed at instant t . This example allows for an instructive comparison of the AARC-based approach with Dynamic Programming, which is the traditional technique for sequential decision-making under dynamical uncertainty. Restricting ourselves with the case where the decision-making problem can be modelled as a Linear Programming problem with the data affected by dynamical uncertainty, we could say that (minimax-oriented) *Dynamic Programming is a specific technique for solving the ARC of this uncertain LP*. Therefore when applicable, Dynamic Programming has a significant advantage as compared to the above AARC-based approach,

since it does *not* impose on the adjustable variables an “ad hoc” restriction (motivated solely by the desire to end up with a tractable problem) to be affine functions of the uncertain data. At the same time, the above “if applicable” is highly restrictive: the computational effort in Dynamical Programming explodes exponentially with the dimension of the state space of the dynamical system in question. For example, the simple Inventory problem we have considered has 4-dimensional state space (the current amount of product in the warehouse plus remaining total capacities of the three factories), which is already computationally too demanding for accurate implementation of Dynamic Programming. The main advantage of the AARC-based dynamical decision-making as compared with Dynamic Programming (as well as with Multi-Stage Stochastic Programming) comes from the “built-in” computational tractability of the approach, which prevents the “curse of dimensionality” and allows to process routinely fairly complicated models with high-dimensional state spaces and many stages.

By the way, it is instructive to compare the AARC approach with Dynamic Programming when the latter is applicable. For example, let us reduce the number of factories in our Inventory problem from 3 to 1, increasing the production capacity of this factory from the previous 567 to 1800 units per period, and let us make the cumulative capacity of the factory equal to 24×1800 , so that the restriction on cumulative production becomes redundant. The resulting dynamical decision-making problem has just one-dimensional state space (all which matters for the future is the current amount of product in the warehouse). Therefore we can easily find by Dynamic Programming the “minimax optimal” inventory management cost (minimum over arbitrary causal⁶⁾ decision rules, maximum over the realizations of the demands from the uncertainty set). With 20% uncertainty, this minimax optimal inventory management cost turns out to be $\text{Opt}_* = 31269.69$. The guarantees for the AARC-based inventory policy can be only worse than for the minimax optimal one: we should pay a price for restricting the decision rules to be affine in the demands. How large is this price? Computation shows that the optimal value in the AARC is $\text{Opt}_{\text{AARC}} = 31514.17$, i.e., it is just by 0.8% larger than the minimax optimal cost Opt_* . And all this – at the uncertainty level as large as 20%! We conclude that the AARC is perhaps not as bad as one could think...

2.5 Does Conic Quadratic Programming exist?

Of course it does. What is meant is whether *SQP exists as an independent entity?*. Specifically, we ask:

(?) Whether a conic quadratic problem can be “efficiently approximated” by a Linear Programming one?

To pose the question formally, let us say that a system of linear inequalities

$$Py + tp + Qu \geq 0 \quad (\text{LP})$$

approximates the conic quadratic inequality

$$\|y\|_2 \leq t \quad (\text{CQI})$$

within accuracy ϵ (or, which is the same, is an ϵ -approximation of (CQI)), if

- (i) Whenever (y, t) satisfies (CQI), there exists u such that (y, t, u) satisfies (LP);
- (ii) Whenever (y, t, u) satisfies (LP), (y, t) “nearly satisfies” (CQI), namely,

$$\|y\|_2 \leq (1 + \epsilon)t. \quad (\text{CQI}_\epsilon)$$

⁶⁾That is, decision of instant t depends solely on the demands at instants $\tau < t$

Note that given a conic quadratic program

$$\min_x \left\{ c^T x : \|A_i x - b_i\|_2 \leq c_i^T x - d_i, i = 1, \dots, m \right\} \quad (\text{CQP})$$

with $m_i \times n$ -matrices A_i and ϵ -approximations

$$P^i y_i + t_i p^i + Q^i u_i \geq 0$$

of conic quadratic inequalities

$$\|y_i\|_2 \leq t_i \quad [\dim y_i = m_i],$$

one can approximate (CQP) by the Linear Programming program

$$\min_{x,u} \left\{ c^T x : P^i(A_i x - b_i) + (c_i^T x - d_i)p^i + Q^i u_i \geq 0, i = 1, \dots, m \right\};$$

if ϵ is small enough, this program, for every practical purpose, is “the same” as (CQP)⁷⁾.

Now, *in principle*, any closed cone of the form

$$\{(y, t) : t \geq \phi(y)\}$$

can be approximated, in the aforementioned sense, by a system of linear inequalities within any accuracy $\epsilon > 0$. The question of crucial importance, however, is how large should be the approximating system – how many linear constraints and additional variables it requires. With naive approach to approximating \mathbf{L}^{n+1} – “take tangent hyperplanes along a fine finite grid of boundary directions and replace the Lorentz cone with the resulting polyhedral one” – the number of linear constraints in, say, 0.5-approximation blows up exponentially as n grows, rapidly making the approximation completely meaningless. Surprisingly, there is a much smarter way to approximate \mathbf{L}^{n+1} :

Theorem 2.5.1 *Let n be the dimension of y in (CQI), and let $0 < \epsilon < 1/2$. There exists (and can be explicitly written) a system of no more than $O(1)n \ln \frac{1}{\epsilon}$ linear inequalities of the form (LP) with $\dim u \leq O(1)n \ln \frac{1}{\epsilon}$ which is an ϵ -approximation of (CQI). Here $O(1)$'s are appropriate absolute constants.*

To get an impression of the constant factors in the Theorem, look at the numbers $I(n, \epsilon)$ of linear inequalities and $V(n, \epsilon)$ of additional variables u in an ϵ -approximation (LP) of the conic quadratic inequality (CQI) with $\dim y = n$:

n	$\epsilon = 10^{-1}$		$\epsilon = 10^{-6}$		$\epsilon = 10^{-14}$	
	$I(n, \epsilon)$	$V(N, \epsilon)$	$I(n, \epsilon)$	$V(n, \epsilon)$	$I(n, \epsilon)$	$V(n, \epsilon)$
4	6	17	31	69	70	148
16	30	83	159	345	361	745
64	133	363	677	1458	1520	3153
256	543	1486	2711	5916	6169	12710
1024	2203	6006	10899	23758	24773	51050

⁷⁾ Note that standard computers do not distinguish between 1 and 1 ± 10^{-17} . It follows that “numerically speaking”, with $\epsilon \sim 10^{-17}$, (CQI) is the same as (CQI $_\epsilon$).

You can see that $I(n, \epsilon) \approx 0.7n \ln \frac{1}{\epsilon}$, $V(n, \epsilon) \approx 2n \ln \frac{1}{\epsilon}$.

The smart approximation described in Theorem 2.5.1 is incomparably better than the outlined naive approximation. On a closest inspection, the “power” of the smart approximation comes from the fact that here we approximate the Lorentz cone by a *projection of a simple higher-dimensional polyhedral cone*. When projecting a polyhedral cone living in \mathbf{R}^N onto a linear subspace of dimension $< N$, you get a polyhedral cone with the number of facets which can be by an exponential in N factor larger than the number of facets of the original cone. Thus, the *projection of a simple (with small number of facets) polyhedral cone onto a subspace of smaller dimension can be a very complicated (with an astronomical number of facets) polyhedral cone*, and this is the fact exploited in the approximation scheme to follow.

2.5.1 Proof of Theorem 2.5.1

Let $\epsilon > 0$ and a positive integer n be given. We intend to build a polyhedral ϵ -approximation of the Lorentz cone \mathbf{L}^{n+1} . Without loss of generality we may assume that n is an integer power of 2: $n = 2^\kappa$, $\kappa \in \mathbf{N}$.

1⁰. “Tower of variables”. The first step of our construction is quite straightforward: we introduce extra variables to represent a conic quadratic constraint

$$\sqrt{y_1^2 + \dots + y_n^2} \leq t \quad (\text{CQI})$$

of dimension $n + 1$ by a system of conic quadratic constraints of dimension 3 each. Namely, let us call our original y -variables “variables of generation 0” and let us split them into pairs $(y_1, y_2), \dots, (y_{n-1}, y_n)$. We associate with every one of these pairs its “successor” – an additional variable “of generation 1”. We split the resulting $2^{\kappa-1}$ variables of generation 1 into pairs and associate with every pair its successor – an additional variable of “generation 2”, and so on; after $\kappa - 1$ steps we end up with two variables of the generation $\kappa - 1$. Finally, the only variable of generation κ is the variable t from (CQI).

To introduce convenient notation, let us denote by y_i^ℓ i -th variable of generation ℓ , so that y_1^0, \dots, y_n^0 are our original y -variables y_1, \dots, y_n , $y_1^\kappa \equiv t$ is the original t -variable, and the “parents” of y_i^ℓ are the variables $y_{2i-1}^{\ell-1}, y_{2i}^{\ell-1}$.

Note that the total number of all variables in the “tower of variables” we end up with is $2n - 1$.

It is clear that the system of constraints

$$\sqrt{[y_{2i-1}^{\ell-1}]^2 + [y_{2i}^{\ell-1}]^2} \leq y_i^\ell, \quad i = 1, \dots, 2^{\kappa-\ell}, \quad \ell = 1, \dots, \kappa \quad (2.5.1)$$

is a representation of (CQI) in the sense that a collection $(y_1^0 \equiv y_1, \dots, y_n^0 \equiv y_n, y_1^\kappa \equiv t)$ can be extended to a solution of (2.5.1) if and only if (y, t) solves (CQI). Moreover, let $\Pi_\ell(x_1, x_2, x_3, u^\ell)$ be polyhedral ϵ_ℓ -approximations of the cone

$$\mathbf{L}^3 = \{(x_1, x_2, x_3) : \sqrt{x_1^2 + x_2^2} \leq x_3\},$$

$\ell = 1, \dots, \kappa$. Consider the system of linear constraints in variables y_i^ℓ, u_i^ℓ :

$$\Pi_\ell(y_{2i-1}^{\ell-1}, y_{2i}^{\ell-1}, y_i^\ell, u_i^\ell) \geq 0, \quad i = 1, \dots, 2^{\kappa-\ell}, \quad \ell = 1, \dots, \kappa. \quad (2.5.2)$$

Writing down this system of linear constraints as $\Pi(y, t, u) \geq 0$, where Π is linear in its arguments, $y = (y_1^0, \dots, y_n^0)$, $t = y_1^\kappa$, and u is the collection of all u_i^ℓ , $\ell = 1, \dots, \kappa$ and all y_i^ℓ , $\ell = 1, \dots, \kappa - 1$, we immediately conclude that Π is a polyhedral ϵ -approximation of \mathbf{L}^{n+1} with

$$1 + \epsilon = \prod_{\ell=1}^{\kappa} (1 + \epsilon_\ell). \quad (2.5.3)$$

In view of this observation, we may focus on building polyhedral approximations of the Lorentz cone \mathbf{L}^3 .

2⁰. Polyhedral approximation of \mathbf{L}^3 we intend to use is given by the system of linear inequalities as follows (positive integer ν is the parameter of the construction):

$$\begin{aligned} (a) \quad & \left\{ \begin{array}{l} \xi^0 \geq |x_1| \\ \eta^0 \geq |x_2| \end{array} \right. \\ (b) \quad & \left\{ \begin{array}{l} \xi^j = \cos\left(\frac{\pi}{2^{j+1}}\right) \xi^{j-1} + \sin\left(\frac{\pi}{2^{j+1}}\right) \eta^{j-1} \\ \eta^j \geq \left| -\sin\left(\frac{\pi}{2^{j+1}}\right) \xi^{j-1} + \cos\left(\frac{\pi}{2^{j+1}}\right) \eta^{j-1} \right|, \quad j = 1, \dots, \nu \end{array} \right. \\ (c) \quad & \left\{ \begin{array}{l} \xi^\nu \leq x_3 \\ \eta^\nu \leq \operatorname{tg}\left(\frac{\pi}{2^{\nu+1}}\right) \xi^\nu \end{array} \right. \end{aligned} \quad (2.5.4)$$

Note that (2.5.4) can be straightforwardly written down as a system of linear homogeneous inequalities $\Pi^{(\nu)}(x_1, x_2, x_3, u) \geq 0$, where u is the collection of $2(\nu+1)$ variables ξ^j, η^j , $j = 0, \dots, \nu$.

Proposition 2.5.1 $\Pi^{(\nu)}$ is a polyhedral $\delta(\nu)$ -approximation of $\mathbf{L}^3 = \{(x_1, x_2, x_3) : \sqrt{x_1^2 + x_2^2} \leq x_3\}$ with

$$\delta(\nu) = \frac{1}{\cos\left(\frac{\pi}{2^{\nu+1}}\right)} - 1. \quad (2.5.5)$$

Proof. We should prove that

- (i) If $(x_1, x_2, x_3) \in \mathbf{L}^3$, then the triple (x_1, x_2, x_3) can be extended to a solution to (2.5.4);
- (ii) If a triple (x_1, x_2, x_3) can be extended to a solution to (2.5.4), then $\|(x_1, x_2)\|_2 \leq (1 + \delta(\nu))x_3$.

(i): Given $(x_1, x_2, x_3) \in \mathbf{L}^3$, let us set $\xi^0 = |x_1|, \eta^0 = |x_2|$, thus ensuring (2.5.4.a). Note that $\|(\xi^0, \eta^0)\|_2 = \|(x_1, x_2)\|_2$ and that the point $P^0 = (\xi^0, \eta^0)$ belongs to the first quadrant.

Now, for $j = 1, \dots, \nu$ let us set

$$\begin{aligned} \xi^j &= \cos\left(\frac{\pi}{2^{j+1}}\right) \xi^{j-1} + \sin\left(\frac{\pi}{2^{j+1}}\right) \eta^{j-1} \\ \eta^j &= \left| -\sin\left(\frac{\pi}{2^{j+1}}\right) \xi^{j-1} + \cos\left(\frac{\pi}{2^{j+1}}\right) \eta^{j-1} \right|, \end{aligned}$$

thus ensuring (2.5.4.b), and let $P^j = (\xi^j, \eta^j)$. The point P^i is obtained from P^{j-1} by the following construction: we rotate clockwise P^{j-1} by the angle $\phi_j = \frac{\pi}{2^{j+1}}$, thus getting a point Q^{j-1} ; if this point is in the upper half-plane, we set $P^j = Q^{j-1}$, otherwise P^j is the reflection of Q^{j-1} with respect to the x -axis. From this description it is clear that

- (I) $\|P^j\|_2 = \|P^{j-1}\|_2$, so that all vectors P^j are of the same Euclidean norm as P^0 , i.e., of the norm $\|(x_1, x_2)\|_2$;

(II) Since the point P^0 is in the first quadrant, the point Q^0 is in the angle $-\frac{\pi}{4} \leq \arg(P) \leq \frac{\pi}{4}$, so that P^1 is in the angle $0 \leq \arg(P) \leq \frac{\pi}{4}$. The latter relation, in turn, implies that Q^1 is in the angle $-\frac{\pi}{8} \leq \arg(P) \leq \frac{\pi}{8}$, whence P^2 is in the angle $0 \leq \arg(P) \leq \frac{\pi}{8}$. Similarly, P^3 is in the angle $0 \leq \arg(P) \leq \frac{\pi}{16}$, and so on: P^j is in the angle $0 \leq \arg(P) \leq \frac{\pi}{2^{j+1}}$.

By (I), $\xi^\nu \leq \|P^\nu\|_2 = \|(x_1, x_2)\|_2 \leq x_3$, so that the first inequality in (2.5.4.c) is satisfied. By (II), P^ν is in the angle $0 \leq \arg(P) \leq \frac{\pi}{2^{\nu+1}}$, so that the second inequality in (2.5.4.c) also is satisfied. We have extended a point from \mathbf{L}^3 to a solution to (2.5.4).

(ii): Let (x_1, x_2, x_3) can be extended to a solution $(x_1, x_2, x_3, \{\xi^j, \eta^j\}_{j=0}^\nu)$ to (2.5.4). Let us set $P^j = (\xi^j, \eta^j)$. From (2.5.4.a, b) it follows that all vectors P^j are nonnegative. We have $\|P^0\|_2 \geq \|(x_1, x_2)\|_2$ by (2.5.4.a). Now, (2.5.4.b) says that the coordinates of P^j are \geq absolute values of the coordinates of P^{j-1} taken in certain orthonormal system of coordinates, so that $\|P^j\|_2 \geq \|P^{j-1}\|_2$. Thus, $\|P^\nu\|_2 \geq \|(x_1, x_2)^T\|_2$. On the other hand, by (2.5.4.c) one has $\|P^\nu\|_2 \leq \frac{1}{\cos(\frac{\pi}{2^{\nu+1}})} \xi^\nu \leq \frac{1}{\cos(\frac{\pi}{2^{\nu+1}})} x_3$, so that $\|(x_1, x_2)^T\|_2 \leq \delta(\nu)x_3$, as claimed. ■

Specifying in (2.5.2) the mappings $\Pi_\ell(\cdot)$ as $\Pi^{(\nu_\ell)}(\cdot)$, we conclude that for every collection of positive integers ν_1, \dots, ν_κ one can point out a polyhedral β -approximation $\Pi_{\nu_1, \dots, \nu_\kappa}(y, t, u)$ of \mathbf{L}^n , $n = 2^\kappa$:

$$\begin{aligned} (a_{\ell,i}) \quad & \left\{ \begin{array}{l} \xi_{\ell,i}^0 \geq |y_{2i-1}| \\ \eta_{\ell,i}^0 \geq |y_{2i}| \end{array} \right. \\ (b_{\ell,i}) \quad & \left\{ \begin{array}{l} \xi_{\ell,i}^j = \cos\left(\frac{\pi}{2^{j+1}}\right) \xi_{\ell,i}^{j-1} + \sin\left(\frac{\pi}{2^{j+1}}\right) \eta_{\ell,i}^{j-1} \\ \eta_{\ell,i}^j \geq \left| -\sin\left(\frac{\pi}{2^{j+1}}\right) \xi_{\ell,i}^{j-1} + \cos\left(\frac{\pi}{2^{j+1}}\right) \eta_{\ell,i}^{j-1} \right|, \quad j = 1, \dots, \nu_\ell \end{array} \right. \\ (c_{\ell,i}) \quad & \left\{ \begin{array}{l} \xi_{\ell,i}^{\nu_\ell} \leq y_i^\ell \\ \eta_{\ell,i}^{\nu_\ell} \leq \operatorname{tg}\left(\frac{\pi}{2^{\nu_\ell+1}}\right) \xi_{\ell,i}^{\nu_\ell} \end{array} \right. \quad i = 1, \dots, 2^{\kappa-\ell}, \quad \ell = 1, \dots, \kappa. \end{aligned} \quad (2.5.6)$$

The approximation possesses the following properties:

1. The dimension of the u -vector (comprised of all variables in (2.5.6) except $y_i = y_i^0$ and $t = y_1^\kappa$) is

$$p(n, \nu_1, \dots, \nu_\kappa) \leq n + O(1) \sum_{\ell=1}^{\kappa} 2^{\kappa-\ell} \nu_\ell;$$

2. The image dimension of $\Pi_{\nu_1, \dots, \nu_\kappa}(\cdot)$ (i.e., the # of linear inequalities plus twice the # of linear equations in (2.5.6)) is

$$q(n, \nu_1, \dots, \nu_\kappa) \leq O(1) \sum_{\ell=1}^{\kappa} 2^{\kappa-\ell} \nu_\ell;$$

3. The quality β of the approximation is

$$\beta = \beta(n; \nu_1, \dots, \nu_\kappa) = \prod_{\ell=1}^{\kappa} \frac{1}{\cos\left(\frac{\pi}{2^{\nu_\ell+1}}\right)} - 1.$$

3⁰. Back to the general case. Given $\epsilon \in (0, 1]$ and setting

$$\nu_\ell = \lfloor O(1)\ell \ln \frac{2}{\epsilon} \rfloor, \quad \ell = 1, \dots, \kappa,$$

with properly chosen absolute constant $O(1)$, we ensure that

$$\begin{aligned} \beta(\nu_1, \dots, \nu_\kappa) &\leq \epsilon, \\ p(n, \nu_1, \dots, \nu_\kappa) &\leq O(1)n \ln \frac{2}{\epsilon}, \\ q(n, \nu_1, \dots, \nu_\kappa) &\leq O(1)n \ln \frac{2}{\epsilon}, \end{aligned}$$

as required. ■

2.6 Exercises

2.6.1 Around randomly perturbed linear constraints

Consider a linear constraint

$$a^T x \geq b \quad [x \in \mathbf{R}^n]. \quad (2.6.1)$$

We have seen that if the coefficients a_j of the left hand side are subject to random perturbations:

$$a_j = a_j^* + \epsilon_j, \quad (2.6.2)$$

where ϵ_j are independent random variables with zero means taking values in segments $[-\sigma_j, \sigma_j]$, then “a reliable version” of the constraint is

$$\sum_j a_j^* x_j - \omega \underbrace{\sqrt{\sum_j \sigma_j^2 x_j^2}}_{\alpha(x)} \geq b, \quad (2.6.3)$$

where $\omega > 0$ is a “safety parameter”. “Reliability” means that if certain x satisfies (2.6.3), then x is “ $\exp\{-\omega^2/4\}$ -reliable solution to (2.6.1)”, that is, the probability that x fails to satisfy a realization of the randomly perturbed constraint (2.6.1) does not exceed $\exp\{-\omega^2/4\}$ (see Proposition 2.4.1). Of course, there exists a possibility to build an “absolutely safe” version of (2.6.1) – (2.6.2) (an analogy of the Robust Counterpart), that is, to require that $\min_{|\epsilon_j| \leq \sigma_j} \sum_j (a_j^* + \epsilon_j)x_j \geq b$, which is exactly the inequality

$$\sum_j a_j^* x_j - \underbrace{\sum_j \sigma_j |x_j|}_{\beta(x)} \geq b. \quad (2.6.4)$$

Whenever x satisfies (2.6.4), x satisfies all realizations of (2.6.1), and not “all, up to exceptions of small probability”. Since (2.6.4) ensures more guarantees than (2.6.3), it is natural to expect from the latter inequality to be “less conservative” than the former one, that is, to expect that the solution set of (2.6.3) is larger than the solution set of (2.6.4). Whether this indeed is the case? The answer depends on the value of the safety parameter ω : when $\omega \leq 1$, the “safety term” $\alpha(x)$ in (2.6.3) is, for every x , not greater than the safety term $\beta(x)$ in (2.6.4), so that every solution to (2.6.4) satisfies (2.6.3). When $\sqrt{n} > \omega > 1$, the “safety terms” in our inequalities become “non-comparable”: depending on x , it may happen that $\alpha(x) \leq \beta(x)$

(which is typical when $\omega \ll \sqrt{n}$), same as it may happen that $\alpha(x) > \beta(x)$. Thus, in the range $1 < \omega < \sqrt{n}$ no one of inequalities (2.6.3), (2.6.4) is more conservative than the other one. Finally, when $\omega \geq \sqrt{n}$, we always have $\alpha(x) \geq \beta(x)$ (why?), so that for “large” values of ω (2.6.3) is even more conservative than (2.6.4). The bottom line is that (2.6.3) is not completely satisfactory candidate to the role of “reliable version” of linear constraint (2.6.1) affected by random perturbations (2.6.2): depending on the safety parameter, this candidate not necessarily is less conservative than the “absolutely reliable” version (2.6.4).

The goal of the subsequent exercises is to build and to investigate an improved version of (2.6.3).

Exercise 2.1 1) Given x , assume that there exist u, v such that

$$\begin{aligned} (a) \quad & x = u + v \\ (b) \quad & \sum_j a_j^* x_j - \sum_j \sigma_j |u_j| - \omega \sqrt{\sum_j \sigma_j^2 v_j^2} \geq b \end{aligned} \quad (2.6.5)$$

Prove that then the probability for x to violate a realization of (2.6.1) is $\leq \exp\{-\omega^2/4\}$ (and is $\leq \exp\{-\omega^2/2\}$ in the case of symmetrically distributed ϵ_j).

2) Verify that the requirement “ x can be extended, by properly chosen u, v , to a solution of (2.6.5)” is weaker than every one of the requirements

- (a) x satisfies (2.6.3)
- (b) x satisfies (2.6.4)

The conclusion of Exercise 2.1 is:

A good “reliable version” of randomly perturbed constraint (2.6.1) – (2.6.2) is system (2.6.5) of linear and conic quadratic constraints in variables x, u, v :

- whenever x can be extended to a solution of system (2.6.5), x is $\exp\{-\omega^2/4\}$ -reliable solution to (2.6.1) (when the perturbations are symmetrically distributed, you can replace $\exp\{-\omega^2/4\}$ with $\exp\{-\omega^2/2\}$);
- at the same time, “as far as x is concerned”, system (2.6.5) is less conservative than every one of the inequalities (2.6.3), (2.6.4): if x solves one of these inequalities, x can be extended to a feasible solution of the system.

Recall that both (2.6.3) and (2.6.4) are Robust Counterparts

$$\min_{a \in \mathcal{U}} a^T x \geq b \quad (2.6.6)$$

of (2.6.1) corresponding to certain choices of the uncertainty set \mathcal{U} : (2.6.3) corresponds to the ellipsoidal uncertainty set

$$\mathcal{U} = \{a : a_j = a_j^* + \sigma_j \zeta_j, \sum_j \zeta_j^2 \leq \omega^2\},$$

while (2.6.3) corresponds to the box uncertainty set

$$\mathcal{U} = \{a : a_j = a_j^* + \sigma_j \zeta_j, \max_j |\zeta_j| \leq 1\}.$$

What about (2.6.5)? Here is the answer:

(!) System (2.6.5) is (equivalent to) the Robust Counterpart (2.6.6) of (2.6.1), the uncertainty set being the intersection of the above ellipsoid and box:

$$\mathcal{U}_* = \{a : a_j = a_j^* + \sigma_j \zeta_j, \sum_j \zeta_j^2 \leq \omega^2, \max_j |\zeta_j| \leq 1\}.$$

Specifically, x can be extended to a feasible solution of (2.6.5) if and only if one has

$$\min_{a \in \mathcal{U}_*} a^T x \geq b.$$

Exercise 2.2 Prove (!) by demonstrating that

$$\max_z \left\{ p^T z : \sum_j z_j^2 \leq R^2, |z_j| \leq 1 \right\} = \min_{u,v} \left\{ \sum_j |u_j| + R \|v\|_2 : u + v = p \right\}.$$

Exercise 2.3 Extend the above constructions and results to the case of uncertain linear inequality

$$a^T x \geq b$$

with certain b and the vector of coefficients a randomly perturbed according to the scheme

$$a = a^* + B\epsilon,$$

where B is deterministic, and the entries $\epsilon_1, \dots, \epsilon_N$ of ϵ are independent random variables with zero means and such that $|\epsilon_i| \leq \sigma_i$ for all i (σ_i are deterministic).

2.6.2 Around Robust Antenna Design

Consider Antenna Design problem as follows:

Given locations $p_1, \dots, p_k \in \mathbf{R}^3$ of k coherent harmonic oscillators, design antenna array which sends as much energy as possible in a given direction (which w.l.o.g. may be taken as the positive direction of the x -axis).

Of course, this is informal setting. The goal of subsequent exercises is to build and process the corresponding model.

Background. In what follows, you can take for granted the following facts:

1. The diagram of “standardly invoked” harmonic oscillator placed at a point $p \in \mathbf{R}^3$ is the following function of a 3D unit direction δ :

$$D_p(\delta) = \cos\left(\frac{2\pi p^T \delta}{\lambda}\right) + i \sin\left(\frac{2\pi p^T \delta}{\lambda}\right) \quad [\delta \in \mathbf{R}^3, \delta^T \delta = 1] \quad (2.6.7)$$

where λ is the wavelength, and i is the imaginary unit.

2. The diagram of an array of oscillators placed at points p_1, \dots, p_k , is the function

$$D(\delta) = \sum_{\ell=1}^k z_\ell D_{p_\ell}(\delta),$$

where z_ℓ are the “element weights” (which form the antenna design and can be arbitrary complex numbers).

3. A natural for engineers way to measure the “concentration” of the energy sent by antenna around a given direction e (which from now on is the positive direction of the x -axis) is
- to choose a $\theta > 0$ and to define the corresponding *sidelobe angle* Δ_θ as the set of all unit 3D directions δ which are at the angle $\geq \theta$ with the direction e ;
 - to measure the “energy concentration” by the *index* $\rho = \frac{|D(e)|}{\max_{\delta \in \Delta_\theta} |D(\delta)|}$, where $D(\cdot)$ is the diagram of the antenna.
4. To make the index easily computable, let us replace in its definition the maximum over the entire sidelobe angle with the maximum over a given “fine finite grid” $\Gamma \subset \Delta_\theta$, thus arriving at the quantity

$$\rho = \frac{|D(e)|}{\max_{\delta \in \Gamma} |D(\delta)|}$$

which we from now on call the *concentration index*.

Developments. Now we can formulate the Antenna Design problem as follows:

(*) Given

- locations p_1, \dots, p_k of harmonic oscillators,
- wavelength λ ,
- finite set Γ of unit 3D directions,

choose complex weights $z_\ell = x_\ell + iy_\ell$, $\ell = 1, \dots, k$ which maximize the index

$$\rho = \frac{|\sum_\ell z_\ell D_\ell(e)|}{\max_{\delta \in \Gamma} |\sum_\ell z_\ell D_\ell(\delta)|} \quad (2.6.8)$$

where $D_\ell(\cdot)$ are given by (2.6.7).

Exercise 2.4 1) Whether the objective (2.6.8) is a concave (and thus “easy to maximize”) function?

2) Prove that (*) is equivalent to the convex optimization program

$$\max_{x_\ell, y_\ell \in \mathbf{R}} \left\{ \Re \left(\sum_\ell (x_\ell + iy_\ell) D_\ell(e) \right) : \left| \sum_\ell (x_\ell + iy_\ell) D_\ell(\delta) \right| \leq 1, \delta \in \Gamma \right\}. \quad (2.6.9)$$

In order to carry out our remaining tasks, it makes sense to approximate (2.6.9) with a Linear Programming problem. To this end, it suffices to approximate the modulus of a complex number z (i.e., the Euclidean norm of a 2D vector) by the quantity

$$\pi_J(z) = \max_{j=1, \dots, J} \Re(\omega_j z) \quad [\omega_j = \cos(\frac{2\pi j}{J}) + i \sin(\frac{2\pi j}{J})]$$

(geometrically: we approximate the unit disk in $\mathbf{C} = \mathbf{R}^2$ by circumscribed perfect J -side polygon).

Exercise 2.5 What is larger – $\pi_J(z)$ or $|z|$? Within which accuracy the “polyhedral norm” $\pi_J(\cdot)$ approximates the modulus?

With the outlined approximation of the modulus, (2.6.9) becomes the optimization program

$$\max_{x_\ell, y_\ell \in \mathbf{R}} \left\{ \Re \left(\sum_\ell (x_\ell + iy_\ell) D_\ell(e) \right) : \Re \left(\omega_j \sum_\ell (x_\ell + iy_\ell) D_\ell(\delta) \right) \leq 1, 1 \leq j \leq J, \delta \in \Gamma \right\}. \quad (2.6.10)$$

Operational Exercise 2.6.1 1) Verify that (2.6.10) is a Linear Programming program and solve it numerically for the following two setups:

Data A:

- $k = 16$ oscillators placed at the points $p_\ell = (\ell - 1)e$, $\ell = 1, \dots, 16$;
- wavelength $\lambda = 2.5$;
- $J = 10$;
- sidelobe grid Γ : since with the oscillators located along the x -axis, the diagram of the array is symmetric with respect to rotations around the x -axis, it suffices to look at the “sidelobe directions” from the xy -plane. To get Γ , we form the set of all directions which are at the angle *at least* $\theta = 0.3\text{rad}$ away from the positive direction of the x -axis, and take 64-point equidistant grid in the resulting “arc of directions”, so that

$$\Gamma = \left\{ \delta_s = \begin{bmatrix} \cos(\alpha + s d\alpha) \\ \sin(\alpha + s d\alpha) \\ 0 \end{bmatrix} \right\}_{s=0}^{63} \quad [\alpha = 0.3, d\alpha = \frac{2(\pi-\alpha)}{63}]$$

Data B: exactly as Data A, except for the wavelength, which is now $\lambda = 5$.

2) Assume that in reality the weights are affected by “implementation errors”:

$$x_\ell = x_\ell^*(1 + \sigma \xi_\ell), \quad y_\ell = y_\ell^*(1 + \sigma \eta_\ell),$$

where x_ℓ^*, y_ℓ^* are the “nominal optimal weights” obtained when solving (2.6.10), x_ℓ, y_ℓ are actual weights, $\sigma > 0$ is the “perturbation level”, and ξ_ℓ, η_ℓ are mutually independent random perturbations uniformly distributed in $[-1, 1]$.

2.1) Check by simulation what happens with the concentration index of the actual diagram as a result of implementation errors. Carry out the simulations for the perturbation level σ taking values 1.e-4, 5.e-4, 1.e-3.

2.2) If you are not satisfied with the behaviour of nominal design(s) in the presence implementation errors, use the Robust Counterpart methodology to replace the nominal designs with the robust ones. What is the “price of robustness” in terms of the index? What do you gain in stability of the diagram w.r.t. implementation errors?

Lecture 3

Semidefinite Programming

In this lecture we study *Semidefinite Programming* – a generic conic program with an extremely wide area of applications.

3.1 Semidefinite cone and Semidefinite programs

3.1.1 Preliminaries

Let \mathbf{S}^m be the space of symmetric $m \times m$ matrices, and $\mathbf{M}^{m,n}$ be the space of rectangular $m \times n$ matrices with real entries. In the sequel, we always think of these spaces as of Euclidean spaces equipped with the *Frobenius* inner product

$$\langle A, B \rangle \equiv \text{Tr}(AB^T) = \sum_{i,j} A_{ij}B_{ij},$$

and we may use in connection with these spaces all notions based upon the Euclidean structure, e.g., the (Frobenius) norm of a matrix

$$\|X\|_2 = \sqrt{\langle X, X \rangle} = \sqrt{\sum_{i,j=1}^m X_{ij}^2} = \sqrt{\text{Tr}(X^T X)}$$

and likewise the notions of orthogonality, orthogonal complement of a linear subspace, etc. Of course, the Frobenius inner product of symmetric matrices can be written without the transposition sign:

$$\langle X, Y \rangle = \text{Tr}(XY), \quad X, Y \in \mathbf{S}^m.$$

Let us focus on the space \mathbf{S}^m . After it is equipped with the Frobenius inner product, we may speak about a cone dual to a given cone $\mathbf{K} \subset \mathbf{S}^m$:

$$\mathbf{K}_* = \{Y \in \mathbf{S}^m \mid \langle Y, X \rangle \geq 0 \quad \forall X \in \mathbf{K}\}.$$

Among the cones in \mathbf{S}^m , the one of special interest is the *semidefinite cone* \mathbf{S}_+^m , the cone of all symmetric positive semidefinite matrices¹⁾. It is easily seen that \mathbf{S}_+^m indeed is a cone, and moreover it is self-dual:

$$(\mathbf{S}_+^m)_* = \mathbf{S}_+^m.$$

¹⁾Recall that a symmetric $n \times n$ matrix A is called positive semidefinite if $x^T Ax \geq 0$ for all $x \in \mathbf{R}^n$; an equivalent definition is that all eigenvalues of A are nonnegative

Another simple fact is that the interior \mathbf{S}_{++}^m of the semidefinite cone \mathbf{S}_+^m is exactly the set of all *positive definite* symmetric $m \times m$ matrices, i.e., symmetric matrices A for which $x^T A x > 0$ for all nonzero vectors x , or, which is the same, symmetric matrices with positive eigenvalues.

The semidefinite cone gives rise to a family of conic programs “minimize a linear objective over the intersection of the semidefinite cone and an affine plane”; these are the *semidefinite programs* we are about to study.

Before writing down a generic semidefinite program, we should resolve a small difficulty with notation. Normally we use lowercase Latin and Greek letters to denote vectors, and the uppercase letters – to denote matrices; e.g., our usual notation for a conic problem is

$$\min_x \left\{ c^T x : Ax - b \geq_{\mathbf{K}} 0 \right\}. \quad (\text{CP})$$

In the case of semidefinite programs, where $\mathbf{K} = \mathbf{S}_+^m$, the usual notation leads to a conflict with the notation related to the space where \mathbf{S}_+^m lives. Look at (CP): without additional remarks it is unclear what is A – is it a $m \times m$ matrix from the space \mathbf{S}^m or is it a linear mapping acting from the space of the design vectors – some \mathbf{R}^n – to the space \mathbf{S}^m ? When speaking about a conic problem on the cone \mathbf{S}_+^m , we should have in mind the second interpretation of A , while the standard notation in (CP) suggests the first (wrong!) interpretation. In other words, we meet with the necessity to distinguish between *linear mappings* acting to/from \mathbf{S}^m and elements of \mathbf{S}^m (which themselves are linear mappings from \mathbf{R}^m to \mathbf{R}^m). In order to resolve the difficulty, we make the following

Notational convention: To denote a linear mapping acting from a linear space to a space of matrices (or from a space of matrices to a linear space), we use uppercase script letters like $\mathcal{A}, \mathcal{B}, \dots$. Elements of usual vector spaces \mathbf{R}^n are, as always, denoted by lowercase Latin/Greek letters $a, b, \dots, z, \alpha, \dots, \zeta$, while elements of a space of matrices usually are denoted by uppercase Latin letters A, B, \dots, Z . According to this convention, a semidefinite program of the form (CP) should be written as

$$\min_x \left\{ c^T x : \mathcal{A}x - B \geq_{\mathbf{S}_+^m} 0 \right\}. \quad (*)$$

We also simplify the sign $\geq_{\mathbf{S}_+^m}$ to \succeq and the sign $>_{\mathbf{S}_+^m}$ to \succ (same as we write \geq instead of $\geq_{\mathbf{R}_+^m}$ and $>$ instead of $>_{\mathbf{R}_+^m}$). Thus, $A \succeq B$ ($\Leftrightarrow B \preceq A$) means that A and B are symmetric matrices of the same size and $A - B$ is positive semidefinite, while $A \succ B$ ($\Leftrightarrow B \prec A$) means that A, B are symmetric matrices of the same size with positive definite $A - B$.

Our last convention is how to write down expressions of the type $A\mathcal{A}xB$ (\mathcal{A} is a linear mapping from some \mathbf{R}^n to \mathbf{S}^m , $x \in \mathbf{R}^n$, $A, B \in \mathbf{S}^m$); what we are trying to denote is the result of the following operation: we first take the value $\mathcal{A}x$ of the mapping \mathcal{A} at a vector x , thus getting an $m \times m$ matrix $\mathcal{A}x$, and then multiply this matrix from the left and from the right by the matrices A, B . In order to avoid misunderstandings, we write expressions of this type as

$$A[\mathcal{A}x]B$$

or as $A\mathcal{A}(x)B$, or as $A\mathcal{A}[x]B$.

How to specify a mapping $\mathcal{A} : \mathbf{R}^n \rightarrow \mathbf{S}^m$. A natural data specifying a linear mapping $A : \mathbf{R}^n \rightarrow \mathbf{R}^m$ is a collection of n elements of the “destination space” – n vectors $a_1, a_2, \dots, a_n \in \mathbf{R}^m$ – such that

$$Ax = \sum_{j=1}^n x_j a_j, \quad x = (x_1, \dots, x_n)^T \in \mathbf{R}^n.$$

Similarly, a natural data specifying a linear mapping $\mathcal{A} : \mathbf{R}^n \rightarrow \mathbf{S}^m$ is a collection A_1, \dots, A_n of n matrices from \mathbf{S}^m such that

$$\mathcal{A}x = \sum_{j=1}^n x_j A_j, \quad x = (x_1, \dots, x_n)^T \in \mathbf{R}^n. \quad (3.1.1)$$

In terms of these data, the semidefinite program (*) can be written as

$$\min_x \left\{ c^T x : x_1 A_1 + x_2 A_2 + \dots + x_n A_n - B \succeq 0 \right\}. \quad (\text{SDPr})$$

It is a simple exercise to verify that if \mathcal{A} is represented as in (3.1.1), then the conjugate to \mathcal{A} linear mapping $\mathcal{A}^* : \mathbf{S}^m \rightarrow \mathbf{R}^n$ is given by

$$\mathcal{A}^* \Lambda = (\text{Tr}(\Lambda A_1), \dots, \text{Tr}(\Lambda A_n))^T : \mathbf{S}^m \rightarrow \mathbf{R}^n. \quad (3.1.2)$$

Linear Matrix Inequality constraints and semidefinite programs. In the case of conic quadratic problems, we started with the simplest program of this type – the one with a single conic quadratic constraint $Ax - b \geq_{\mathbf{L}^m} 0$ – and then defined a conic quadratic program as a program with finitely many constraints of this type, i.e., as a conic program on a *direct product* of the ice-cream cones. In contrast to this, when defining a semidefinite program, we impose on the design vector just one Linear Matrix Inequality (LMI) $\mathcal{A}x - B \succeq 0$. Now we indeed should not bother about more than a single LMI, due to the following simple fact:

A system of finitely many LMI's

$$\mathcal{A}_i x - B_i \succeq 0, \quad i = 1, \dots, k,$$

is equivalent to the single LMI

$$\mathcal{A}x - B \succeq 0,$$

with

$$\mathcal{A}x = \text{Diag}(\mathcal{A}_1 x, \mathcal{A}_2 x, \dots, \mathcal{A}_k x), \quad B = \text{Diag}(B_1, \dots, B_k);$$

here for a collection of symmetric matrices Q_1, \dots, Q_k

$$\text{Diag}(Q_1, \dots, Q_k) = \begin{pmatrix} Q_1 & & \\ & \ddots & \\ & & Q_k \end{pmatrix}$$

is the block-diagonal matrix with the diagonal blocks Q_1, \dots, Q_k .

Indeed, a block-diagonal symmetric matrix is positive (semi)definite if and only if all its diagonal blocks are so.

3.1.1.1 Dual to a semidefinite program (SDP)

Specifying the general concept of conic dual of a conic program in the case when the latter is a semidefinite program (*) and taking into account (3.1.2) along with the fact that the semidefinite cone is self-dual, we see that the dual to (*) is the semidefinite program

$$\max_{\Lambda} \{ \langle B, \Lambda \rangle \equiv \text{Tr}(B\Lambda) : \text{Tr}(A_i \Lambda) = c_i, \quad i = 1, \dots, n; \quad \Lambda \succeq 0 \}. \quad (\text{SDDl})$$

3.1.1.2 Conic Duality in the case of Semidefinite Programming

Let us see what we get from the Conic Duality Theorem in the case of semidefinite programs. First note that our default assumption **A** on a conic program in the form of (CP) (Lecture 1) as applied to (SDPr) says that no nontrivial linear combination of the matrices A_1, \dots, A_n is 0. Strict feasibility of (SDPr) means that there exists x such that $\mathcal{A}x - B$ is positive definite, and strict feasibility of (SDDl) means that there exists a positive definite Λ satisfying $\mathcal{A}^*\Lambda = c$. According to the Conic Duality Theorem, if both primal and dual are strictly feasible, both are solvable, the optimal values are equal to each other, and the complementary slackness condition

$$[\text{Tr}(\Lambda[\mathcal{A}x - B]) \equiv] \quad \langle \Lambda, \mathcal{A}x - B \rangle = 0$$

is necessary and sufficient for a pair of a primal feasible solution x and a dual feasible solution Λ to be optimal for the corresponding problems.

It is easily seen that for a pair X, Y of positive semidefinite symmetric matrices one has

$$\text{Tr}(XY) = 0 \Leftrightarrow XY = YX = 0;$$

in particular, in the case of strictly feasible primal and dual problems, the “primal slack” $S_* = \mathcal{A}x^* - B$ corresponding to a primal optimal solution commutes with (any) dual optimal solution Λ_* , and the product of these two matrices is 0. Besides this, S_* and Λ_* , as a pair of commuting symmetric matrices, share a common eigenbasis, and the fact that $S_*\Lambda_* = 0$ means that the eigenvalues of the matrices in this basis are “complementary”: for every common eigenvector, either the eigenvalue of S_* , or the one of Λ_* , or both, are equal to 0 (cf. with complementary slackness in the LP case).

3.2 What can be expressed via LMI's?

As in the previous lecture, the first thing to realize when speaking about the “semidefinite programming universe” is how to recognize that a convex optimization program

$$\min_x \left\{ c^T x : x \in X = \bigcap_{i=1}^m X_i \right\} \tag{P}$$

can be cast as a semidefinite program. Just as in the previous lecture, this question actually asks whether a given convex set/convex function is positive semidefinite representable (in short: SDr). The definition of the latter notion is completely similar to the one of a CQr set/function:

We say that a convex set $X \subset \mathbf{R}^n$ is SDr, if there exists an affine mapping $(x, u) \rightarrow \mathcal{A} \begin{pmatrix} x \\ u \end{pmatrix} - B : \mathbf{R}_x^n \times \mathbf{R}_u^k \rightarrow \mathbf{S}^m$ such that

$$x \in X \Leftrightarrow \exists u : \mathcal{A} \begin{pmatrix} x \\ u \end{pmatrix} - B \succeq 0;$$

in other words, X is SDr, if there exists LMI

$$\mathcal{A} \begin{pmatrix} x \\ u \end{pmatrix} - B \succeq 0,$$

in the original design vector x and a vector u of additional design variables such that X is a projection of the solution set of the LMI onto the x -space. An LMI with this property is called Semidefinite Representation (SDR) of the set X .

A convex function $f : \mathbf{R}^n \rightarrow \mathbf{R} \cup \{+\infty\}$ is called SDR, if its epigraph

$$\{(x, t) \mid t \geq f(x)\}$$

is a SDR set. A SDR of the epigraph of f is called semidefinite representation of f .

By exactly the same reasons as in the case of conic quadratic problems, one has:

1. If f is a SDR function, then all its level sets $\{x \mid f(x) \leq a\}$ are SDR; the SDR of the level sets are explicitly given by (any) SDR of f ;
2. If all the sets X_i in problem (P) are SDR with known SDR's, then the problem can explicitly be converted to a semidefinite program.

In order to understand which functions/sets are SDR, we may use the same approach as in Lecture 2. “The calculus”, i.e., the list of basic operations preserving SD-representability, is exactly the same as in the case of conic quadratic problems; we just may repeat word by word the relevant reasoning from Lecture 2, each time replacing “CQr” with “SDR”. Thus, the only issue to be addressed is the derivation of a catalogue of “simple” SDR functions/sets. Our first observation in this direction is as follows:

1-17. ²⁾ If a function/set is CQr, it is also SDR, and any CQR of the function/set can be explicitly converted to its SDR.

Indeed, the notion of a CQr/SDR function is a “derivative” of the notion of a CQr/SDR set: by definition, a function is CQr/SDR if and only if its epigraph is so. Now, CQr sets are exactly those sets which can be obtained as projections of the solution sets of systems of conic quadratic inequalities, i.e., as projections of inverse images, under affine mappings, of direct products of ice-cream cones. Similarly, SDR sets are projections of the inverse images, under affine mappings, of positive semidefinite cones. Consequently,

- (i) in order to verify that a CQr set is SDR as well, it suffices to show that an inverse image, under an affine mapping, of a direct product of ice-cream cones – a set of the form

$$Z = \{z \mid Az - b \in \mathbf{K} = \prod_{i=1}^l \mathbf{L}^{k_i}\}$$

is the inverse image of a semidefinite cone under an affine mapping. To this end, in turn, it suffices to demonstrate that

- (ii) a direct product $\mathbf{K} = \prod_{i=1}^l \mathbf{L}^{k_i}$ of ice-cream cones is an inverse image of a semidefinite cone under an affine mapping.

Indeed, representing \mathbf{K} as $\{y \mid \mathcal{A}y - b \in \mathbf{S}_+^m\}$, we get

$$Z = \{z \mid Az - b \in \mathbf{K}\} = \{z \mid \hat{\mathcal{A}}z - \hat{b} \in \mathbf{S}_+^m\},$$

²⁾We refer to Examples 1-17 of CQ-representable functions/sets from Section 2.3

where $\hat{A}z - \hat{B} = \mathcal{A}(Az - b) - B$ is affine.

In turn, in order to prove (ii) it suffices to show that

(iii) Every ice-cream cone \mathbf{L}^k is an inverse image of a semidefinite cone under an affine mapping.

In fact the implication (iii) \Rightarrow (ii) is given by our calculus, since a direct product of SDr sets is again SDr³⁾.

We have reached the point where no more reductions are necessary, and here is the demonstration of (iii). To see that the Lorentz cone \mathbf{L}^k , $k > 1$, is SDr, it suffices to observe that

$$\begin{pmatrix} x \\ t \end{pmatrix} \in \mathbf{L}^k \Leftrightarrow \mathcal{A}(x, t) = \begin{pmatrix} tI_{k-1} & x \\ x^T & t \end{pmatrix} \succeq 0 \quad (3.2.1)$$

(x is $k-1$ -dimensional, t is scalar, I_{k-1} is the $(k-1) \times (k-1)$ unit matrix). (3.2.1) indeed resolves the problem, since the matrix $\mathcal{A}(x, t)$ is linear in (x, t) !

It remains to verify (3.2.1), which is immediate. If $(x, t) \in \mathbf{L}^k$, i.e., if $\|x\|_2 \leq t$, then for every $y = \begin{pmatrix} \xi \\ \tau \end{pmatrix} \in \mathbf{R}^k$ (ξ is $(k-1)$ -dimensional, τ is scalar) we have

$$\begin{aligned} y^T \mathcal{A}(x, t) y &= \tau^2 t + 2\tau x^T \xi + t \xi^T \xi \geq \tau^2 t - 2|\tau| \|x\|_2 \|\xi\|_2 + t \|\xi\|_2^2 \\ &\geq t\tau^2 - 2t|\tau| \|\xi\|_2 + t \|\xi\|_2^2 \\ &\geq t(|\tau| - \|\xi\|_2)^2 \geq 0, \end{aligned}$$

so that $\mathcal{A}(x, t) \succeq 0$. Vice versa, if $\mathcal{A}(t, x) \succeq 0$, then of course $t \geq 0$. Assuming $t = 0$, we immediately obtain $x = 0$ (since otherwise for $y = \begin{pmatrix} x \\ 0 \end{pmatrix}$ we would have $0 \leq y^T \mathcal{A}(x, t) y = -2\|x\|_2^2$); thus, $\mathcal{A}(x, t) \succeq 0$ implies $\|x\|_2 \leq t$ in the case of $t = 0$. To see that the same implication is valid for $t > 0$, let us set $y = \begin{pmatrix} -x \\ t \end{pmatrix}$ to get

$$0 \leq y^T \mathcal{A}(x, t) y = tx^T x - 2tx^T x + t^3 = t(t^2 - x^T x),$$

whence $\|x\|_2 \leq t$, as claimed. ■

We see that the “expressive abilities” of semidefinite programming are even richer than those of Conic Quadratic programming. In fact the gap is quite significant. The first new possibility is the ability to handle eigenvalues, and the importance of this possibility can hardly be overestimated.

3.2.0.3 SD-representability of functions of eigenvalues of symmetric matrices

Our first eigenvalue-related observation is as follows:

³⁾ Just to recall where the calculus comes from, here is a direct verification:

Given a direct product $\mathbf{K} = \prod_{i=1}^l \mathbf{L}^{k_i}$ of ice-cream cones and given that every factor in the product is the inverse image of a semidefinite cone under an affine mapping:

$$\mathbf{L}^{k_i} = \{x_i \in \mathbf{R}^{k_i} \mid \mathcal{A}_i x_i - B_i \succeq 0\},$$

we can represent \mathbf{K} as the inverse image of a semidefinite cone under an affine mapping, namely, as

$$\mathbf{K} = \{x = (x_1, \dots, x_l) \in \mathbf{R}^{k_1} \times \dots \times \mathbf{R}^{k_l} \mid \text{Diag}(\mathcal{A}_1 x_1 - B_1, \dots, \mathcal{A}_l x_l - B_l) \succeq 0\}.$$

18. The largest eigenvalue $\lambda_{\max}(X)$ regarded as a function of $m \times m$ symmetric matrix X is SDr. Indeed, the epigraph of this function

$$\{(X, t) \in \mathbf{S}^m \times \mathbf{R} \mid \lambda_{\max}(X) \leq t\}$$

is given by the LMI

$$tI_m - X \succeq 0,$$

where I_m is the unit $m \times m$ matrix.

Indeed, the eigenvalues of $tI_m - X$ are t minus the eigenvalues of X , so that the matrix $tI_m - X$ is positive semidefinite – all its eigenvalues are nonnegative – if and only if t majorates all eigenvalues of X .

The latter example admits a natural generalization. Let M, A be two symmetric $m \times m$ matrices, and let M be positive definite. A real λ and a nonzero vector e are called eigenvalue and eigenvector of the pencil $[M, A]$, if $Ae = \lambda Me$ (in particular, the usual eigenvalues/eigenvectors of A are exactly the eigenvalues/eigenvectors of the pencil $[I_m, A]$). Clearly, λ is an eigenvalue of $[M, A]$ if and only if the matrix $\lambda M - A$ is singular, and nonzero vectors from the kernel of the latter matrix are exactly the eigenvectors of $[M, A]$ associated with the eigenvalue λ . The eigenvalues of the pencil $[M, A]$ are the usual eigenvalues of the matrix $M^{-1/2}AM^{-1/2}$, as can be concluded from:

$$\text{Det}(\lambda M - A) = 0 \Leftrightarrow \text{Det}(M^{1/2}(\lambda I_m - M^{-1/2}AM^{-1/2})M^{1/2}) = 0 \Leftrightarrow \text{Det}(\lambda I_m - M^{-1/2}AM^{-1/2}) = 0.$$

The announced extension of Example 18 is as follows:

18a. [The maximum eigenvalue of a pencil]: Let M be a positive definite symmetric $m \times m$ matrix, and let $\lambda_{\max}(X : M)$ be the largest eigenvalue of the pencil $[M, X]$, where X is a symmetric $m \times m$ matrix. The inequality

$$\lambda_{\max}(X : M) \leq t$$

is equivalent to the matrix inequality

$$tM - X \succeq 0.$$

In particular, $\lambda_{\max}(X : M)$, regarded as a function of X , is SDr.

18b. The spectral norm $|X|$ of a symmetric $m \times m$ matrix X , i.e., the maximum of absolute values of the eigenvalues of X , is SDr. Indeed, a SDR of the epigraph

$$\{(X, t) \mid |X| \leq t\} = \{(X, t) \mid \lambda_{\max}(X) \leq t, \lambda_{\max}(-X) \leq t\}$$

of $|X|$ is given by the pair of LMI's

$$tI_m - X \succeq 0, tI_m + X \succeq 0.$$

In spite of their simplicity, the indicated results are extremely useful. As a more complicated example, let us build a SDr for the sum of the k largest eigenvalues of a symmetric matrix.

From now on, speaking about $m \times m$ symmetric matrix X , we denote by $\lambda_i(X)$, $i = 1, \dots, m$, its eigenvalues *counted with their multiplicities and arranged in a non-ascending order*:

$$\lambda_1(X) \geq \lambda_2(X) \geq \dots \geq \lambda_m(X).$$

The vector of the eigenvalues (in the indicated order) will be denoted $\lambda(X)$:

$$\lambda(X) = (\lambda_1(X), \dots, \lambda_m(X))^T \in \mathbf{R}^m.$$

The question we are about to address is which functions of the eigenvalues are SDr. We already know that this is the case for the largest eigenvalue $\lambda_1(X)$. Other eigenvalues cannot be SDr since they are not convex functions of X . And convexity, of course, is a necessary condition for SD-representability (cf. Lecture 2). It turns out, however, that the m functions

$$S_k(X) = \sum_{i=1}^k \lambda_i(X), \quad k = 1, \dots, m,$$

are convex and, moreover, are SDr:

18c. Sums of largest eigenvalues of a symmetric matrix. Let X be $m \times m$ symmetric matrix, and let $k \leq m$. Then the function $S_k(X)$ is SDr. Specifically, the epigraph

$$\{(X, t) \mid S_k(x) \leq t\}$$

of the function admits the SDR

$$\begin{aligned} (a) \quad & t - ks - \text{Tr}(Z) \geq 0 \\ (b) \quad & Z \succeq 0 \\ (c) \quad & Z - X + sI_m \succeq 0 \end{aligned} \tag{3.2.2}$$

where $Z \in \mathbf{S}^m$ and $s \in \mathbf{R}$ are additional variables.

We should prove that

- (i) If a given pair X, t can be extended, by properly chosen s, Z , to a solution of the system of LMI's (3.2.2), then $S_k(X) \leq t$;
- (ii) Vice versa, if $S_k(X) \leq t$, then the pair X, t can be extended, by properly chosen s, Z , to a solution of (3.2.2).

To prove (i), we use the following basic fact⁴⁾:

(W) The vector $\lambda(X)$ is a \succeq -monotone function of $X \in \mathbf{S}^m$:

$$X \succeq X' \Rightarrow \lambda(X) \geq \lambda(X').$$

Assuming that (X, t, s, Z) is a solution to (3.2.2), we get $X \preceq Z + sI_m$, so that

$$\lambda(X) \leq \lambda(Z + sI_m) = \lambda(Z) + s \begin{pmatrix} 1 \\ \vdots \\ 1 \end{pmatrix},$$

whence

$$S_k(X) \leq S_k(Z) + sk.$$

⁴⁾ which is an immediate corollary of the fundamental *Variational Characterization of Eigenvalues* of symmetric matrices, see Section A.7.3: for a symmetric $m \times m$ matrix A ,

$$\lambda_i(A) = \min_{E \in \mathcal{E}_i} \max_{e \in E: e^T e = 1} e^T A e,$$

where \mathcal{E}_i is the collection of all linear subspaces of the dimension $n - i + 1$ in \mathbf{R}^m ,

Since $Z \succeq 0$ (see (3.2.2.b)), we have $S_k(Z) \leq \text{Tr}(Z)$, and combining these inequalities we get

$$S_k(X) \leq \text{Tr}(Z) + sk.$$

The latter inequality, in view of (3.2.2.a)), implies $S_k(X) \leq t$, and (i) is proved.

To prove (ii), assume that we are given X, t with $S_k(X) \leq t$, and let us set $s = \lambda_k(X)$. Then the k largest eigenvalues of the matrix $X - sI_m$ are nonnegative, and the remaining are nonpositive. Let Z be a symmetric matrix with the same eigenbasis as X and such that the k largest eigenvalues of Z are the same as those of $X - sI_m$, and the remaining eigenvalues are zeros. The matrices Z and $Z - X + sI_m$ are clearly positive semidefinite (the first by construction, and the second since in the eigenbasis of X this matrix is diagonal with the first k diagonal entries being 0 and the remaining being the same as those of the matrix $sI_m - X$, i.e., nonnegative). Thus, the matrix Z and the real s we have built satisfy (3.2.2.b, c). In order to see that (3.2.2.a) is satisfied as well, note that by construction $\text{Tr}(Z) = S_k(x) - sk$, whence $t - sk - \text{Tr}(Z) = t - S_k(x) \geq 0$.

In order to proceed, we need the following highly useful technical result:

Lemma 3.2.1 [Lemma on the Schur Complement] *Let*

$$A = \begin{pmatrix} B & C^T \\ C & D \end{pmatrix}$$

be a symmetric matrix with $k \times k$ block B and $\ell \times \ell$ block D . Assume that B is positive definite. Then A is positive (semi)definite if and only if the matrix

$$D - CB^{-1}C^T$$

is positive (semi)definite (this matrix is called the Schur complement of B in A).

Proof. The positive semidefiniteness of A is equivalent to the fact that

$$0 \leq (x^T, y^T) \begin{pmatrix} B & C^T \\ C & D \end{pmatrix} \begin{pmatrix} x \\ y \end{pmatrix} = x^T B x + 2x^T C^T y + y^T D y \quad \forall x \in \mathbf{R}^k, y \in \mathbf{R}^\ell,$$

or, which is the same, to the fact that

$$\inf_{x \in \mathbf{R}^k} [x^T B x + 2x^T C^T y + y^T D y] \geq 0 \quad \forall y \in \mathbf{R}^\ell.$$

Since B is positive definite by assumption, the infimum in x can be computed explicitly for every fixed y : the optimal x is $-B^{-1}C^T y$, and the optimal value is

$$y^T D y - y^T C B^{-1} C^T y = y^T [D - CB^{-1}C^T] y.$$

The positive definiteness/semidefiniteness of A is equivalent to the fact that the latter expression is, respectively, positive/nonnegative for every $y \neq 0$, i.e., to the positive definiteness/semidefiniteness of the Schur complement of B in A . ■

18d. “Determinant” of a symmetric positive semidefinite matrix. Let X be a symmetric positive semidefinite $m \times m$ matrix. Although its determinant

$$\text{Det}(X) = \prod_{i=1}^m \lambda_i(X)$$

is neither a convex nor a concave function of X (if $m \geq 2$), it turns out that the function $\text{Det}^q(X)$ is concave in X whenever $0 \leq q \leq \frac{1}{m}$. Functions of this type are important in many volume-related problems (see below); we are about to prove that

if q is a rational number, $0 \leq q \leq \frac{1}{m}$, then the function

$$f_q(X) = \begin{cases} -\text{Det}^q(X), & X \succeq 0 \\ +\infty, & \text{otherwise} \end{cases}$$

is SDR.

Consider the following system of LMI's:

$$\begin{pmatrix} X & \Delta \\ \Delta^T & D(\Delta) \end{pmatrix} \succeq 0, \quad (\text{D})$$

where Δ is $m \times m$ lower triangular matrix comprised of additional variables, and $D(\Delta)$ is the diagonal matrix with the same diagonal entries as those of Δ . Let $\text{diag}(\Delta)$ denote the vector of the diagonal entries of the square matrix Δ .

As we know from Lecture 2 (see Example 15), the set

$$\{(\delta, t) \in \mathbf{R}_+^m \times \mathbf{R} \mid t \leq (\delta_1 \dots \delta_m)^q\}$$

admits an explicit CQR. Consequently, this set admits an explicit SDR as well. The latter SDR is given by certain LMI $S(\delta, t; u) \succeq 0$, where u is the vector of additional variables of the SDR, and $S(\delta, t, u)$ is a matrix affinely depending on the arguments. We claim that

(!) The system of LMI's (D) & $S(\text{diag}(\Delta), t; u) \succeq 0$ is a SDR for the set

$$\{(X, t) \mid X \succeq 0, t \leq \text{Det}^q(X)\},$$

which is basically the epigraph of the function f_q (the latter is obtained from our set by reflection with respect to the plane $t = 0$).

To support our claim, recall that by Linear Algebra a matrix X is positive semidefinite if and only if it can be factorized as $X = \widehat{\Delta} \widehat{\Delta}^T$ with a lower triangular $\widehat{\Delta}$, $\text{diag}(\widehat{\Delta}) \geq 0$; the resulting matrix $\widehat{\Delta}$ is called the Choleski factor of X . Note that if $X \succeq 0$ and $t \leq \text{Det}^q(X)$, then

(1) We can extend X by appropriately chosen lower triangular matrix Δ to a solution of (D) in such a way that if $\delta = \text{diag}(\Delta)$, then $\prod_{i=1}^m \delta_i = \text{Det}(X)$.

Indeed, let $\widehat{\Delta}$ be the Choleski factor of X . Let \widehat{D} be the diagonal matrix with the same diagonal entries as those of $\widehat{\Delta}$, and let $\Delta = \widehat{\Delta} \widehat{D}$, so that the diagonal entries δ_i of Δ are squares of the diagonal entries $\widehat{\delta}_i$ of the matrix $\widehat{\Delta}$. Thus, $D(\Delta) = \widehat{D}^2$. It follows that for every $\epsilon > 0$ one has $\Delta[D(\Delta) + \epsilon I]^{-1} \Delta^T = \widehat{\Delta} \widehat{D} [\widehat{D}^2 + \epsilon I]^{-1} \widehat{D} \widehat{\Delta}^T \preceq \widehat{\Delta} \widehat{\Delta}^T = X$. We

see that by the Schur Complement Lemma all matrices of the form $\begin{pmatrix} X & \Delta \\ \Delta^T & D(\Delta) + \epsilon I \end{pmatrix}$

with $\epsilon > 0$ are positive semidefinite, whence $\begin{pmatrix} X & \Delta \\ \Delta^T & D(\Delta) \end{pmatrix} \succeq 0$. Thus, (D) is indeed

satisfied by (X, Δ) . And of course $X = \widehat{\Delta} \widehat{\Delta}^T \Rightarrow \text{Det}(X) = \text{Det}^2(\widehat{\Delta}) = \prod_{i=1}^m \widehat{\delta}_i^2 = \prod_{i=1}^m \delta_i$.

(2) Since $\delta = \text{diag}(\Delta) \geq 0$ and $\prod_{i=1}^m \delta_i = \text{Det}(X)$, we get $t \leq \text{Det}^q(X) = \left(\prod_{i=1}^m \delta_i \right)^q$, so that we can extend (t, δ) by a properly chosen u to a solution of the LMI $S(\text{diag}(\Delta), t; u) \succeq 0$.

We conclude that if $X \succeq 0$ and $t \leq \text{Det}^q(X)$, then one can extend the pair X, t by properly chosen Δ and u to a solution of the LMI (D) & $S(\text{diag}(\Delta), t; u) \succeq 0$, which is the first part of the proof of (!).

To complete the proof of (!), it suffices to demonstrate that if for a given pair X, t there exist Δ and u such that (D) and the LMI $S(\text{diag}(\Delta), t; u) \succeq 0$ are satisfied, then X is positive semidefinite and $t \leq \text{Det}^q(X)$. This is immediate: denoting $\delta = \text{diag}(\Delta) [\geq 0]$ and applying the Schur Complement Lemma, we conclude that $X \succeq \Delta[D(\Delta) + \epsilon I]^{-1}\Delta^T$ for every $\epsilon > 0$. Applying (W), we get $\lambda(X) \geq \lambda(\Delta[D(\Delta) + \epsilon I]^{-1}\Delta^T)$, whence of course $\text{Det}(X) \geq \text{Det}(\Delta[D(\Delta) + \epsilon I]^{-1}\Delta^T) = \prod_{i=1}^m \delta_i^2 / (\delta_i + \epsilon)$. Passing to limit as $\epsilon \rightarrow 0$, we get $\prod_{i=1}^m \delta_i \leq \text{Det}(X)$. On the other hand, the LMI $S(\delta, t; u) \succeq 0$ takes place, which means that $t \leq \left(\prod_{i=1}^m \delta_i \right)^q$. Combining the resulting inequalities, we come to $t \leq \text{Det}^q(X)$, as required. ■

18e. Negative powers of the determinant. Let q be a positive rational. Then the function

$$f(X) = \begin{cases} \text{Det}^{-q}(X), & X \succ 0 \\ +\infty, & \text{otherwise} \end{cases}$$

of symmetric $m \times m$ matrix X is SDr.

The construction is completely similar to the one used in Example 18d. As we remember from Lecture 2, Example 16, the function $g(\delta) = (\delta_1 \dots \delta_m)^{-q}$ of positive vector $\delta = (\delta_1, \dots, \delta_m)^T$ is CQr and is therefore SDr as well. Let an SDR of the function be given by LMI $\mathcal{R}(\delta, t, u) \succeq 0$. The same arguments as in Example 18d demonstrate that the pair of LMI's (D) & $\mathcal{R}(\text{Dg}(\Delta), t, u) \succeq 0$ is an SDR for f .

In examples 18, 18b – 18d we were discussed SD-representability of particular functions of eigenvalues of a symmetric matrix. Here is a general statement of this type:

Proposition 3.2.1 *Let $g(x_1, \dots, x_m) : \mathbf{R}^m \rightarrow \mathbf{R} \cup \{+\infty\}$ be a symmetric (i.e., invariant with respect to permutations of the coordinates x_1, \dots, x_m) SD-representable function:*

$$t \geq f(x) \Leftrightarrow \exists u : \mathcal{S}(x, t, u) \succeq 0,$$

with \mathcal{S} affinely depending on x, t, u . Then the function

$$f(X) = g(\lambda(X))$$

of symmetric $m \times m$ matrix X is SDr, with SDR given by the relation

$$(a) \quad t \geq f(X) \quad \Updownarrow$$

$$\exists x_1, \dots, x_m, u : \quad \begin{aligned} (b) \quad & \left\{ \begin{array}{l} \mathcal{S}(x_1, \dots, x_m, t, u) \succeq 0 \\ x_1 \geq x_2 \geq \dots \geq x_m \\ S_j(X) \leq x_1 + \dots + x_j, \quad j = 1, \dots, m-1 \\ \text{Tr}(X) = x_1 + \dots + x_m \end{array} \right. \end{aligned} \quad (3.2.3)$$

(recall that the functions $S_j(X) = \sum_{i=1}^k \lambda_i(X)$ are SDr, see Example 18c). Thus, the solution set of (b) is SDr (as an intersection of SDr sets), which implies SD-representability of the projection of this set onto the (X, t) -plane; by (3.2.3) the latter projection is exactly the epigraph of f).

The proof of Proposition 3.2.1 is based upon an extremely useful result known as *Birkhoff's Theorem*⁵⁾.

As a corollary of Proposition 3.2.1, we see that the following functions of a symmetric $m \times m$ matrix X are SDR:

- $f(X) = -\text{Det}^q(X)$, $X \succeq 0$, $q \leq \frac{1}{m}$ is a positive rational (this fact was already established directly);
 [here $g(x_1, \dots, x_m) = (x_1 \dots x_m)^q : \mathbf{R}_+^n \rightarrow \mathbf{R}$; a CQR (and thus – a SDR) of g is presented in Example 15 of Lecture 2]
- $f(x) = \text{Det}^{-q}(X)$, $X \succ 0$, q is a positive rational (cf. Example 18e)
 [here $g(x_1, \dots, x_m) = (x_1, \dots, x_m)^{-q} : \mathbf{R}_{++}^m \rightarrow \mathbf{R}$; a CQR of g is presented in Example 16 of Lecture 2]
- $\|X\|_p = \left(\sum_{i=1}^m |\lambda_i(X)|^p \right)^{1/p}$, $p \geq 1$ is rational
 [$g(x) = \|x\|_p \equiv \left(\sum_{i=1}^m |x_i|^p \right)^{1/p}$, see Lecture 2, Example 17a]
- $\|X_+\|_p = \left(\sum_{i=1}^m \max^p[\lambda_i(X), 0] \right)^{1/p}$, $p \geq 1$ is rational
 [$g(x) = \|x_+\|_p \equiv \left(\sum_{i=1}^m |\max^p[x_i, 0]| \right)^{1/p}$, see Lecture 2, Example 17b]

SD-representability of functions of singular values. Consider the space $\mathbf{M}^{k,l}$ of $k \times l$ rectangular matrices and assume that $k \leq l$. Given a matrix $A \in \mathbf{M}^{k,l}$, consider the symmetric positive semidefinite $k \times k$ matrix $(AA^T)^{1/2}$; its eigenvalues are called *singular values* of A and are denoted by $\sigma_1(A), \dots, \sigma_k(A)$: $\sigma_i(A) = \lambda_i((AA^T)^{1/2})$. According to the convention on how we enumerate eigenvalues of a symmetric matrix, the singular values form a non-ascending sequence:

$$\sigma_1(A) \geq \sigma_2(A) \geq \dots \geq \sigma_k(A).$$

The importance of the singular values comes from the Singular Value Decomposition Theorem which states that a $k \times l$ matrix A ($k \leq l$) can be represented as

$$A = \sum_{i=1}^k \sigma_i(A) e_i f_i^T,$$

⁵⁾The Birkhoff Theorem, which, aside of other applications, implies a number of crucial facts about eigenvalues of symmetric matrices, by itself even does not mention the word “eigenvalue” and reads: *The extreme points of the polytope \mathcal{P} of double stochastic $m \times m$ matrices – those with nonnegative entries and unit sums of entries in every row and every column – are exactly the permutation matrices (those with a single nonzero entry, equal to 1, in every row and every column).* For proof, see Section B.2.8.C.1.

where $\{e_i\}_{i=1}^k$ and $\{f_i\}_{i=1}^k$ are orthonormal sequences in \mathbf{R}^k and \mathbf{R}^l , respectively; this is a surrogate of the eigenvalue decomposition of a symmetric $k \times k$ matrix

$$A = \sum_{i=1}^k \lambda_i(A) e_i e_i^T,$$

where $\{e_i\}_{i=1}^k$ form an orthonormal eigenbasis of A .

Among the singular values of a rectangular matrix, the most important is the largest $\sigma_1(A)$. This is nothing but the *operator (or spectral) norm* of A :

$$|A| = \max\{\|Ax\|_2 \mid \|x\|_2 \leq 1\}.$$

For a symmetric matrix, the singular values are exactly the modulae of the eigenvalues, and our new definition of the norm coincides with the one already given in 18b.

It turns out that the sum of a given number of the largest singular values of A

$$\Sigma_p(A) = \sum_{i=1}^p \sigma_i(A)$$

is a convex and, moreover, a SDr function of A . In particular, the operator norm of A is SDr:

19. The sum $\Sigma_p(X)$ of p largest singular values of a rectangular matrix $X \in \mathbf{M}^{k,l}$ is SDr. In particular, the operator norm of a rectangular matrix is SDr:

$$|X| \leq t \Leftrightarrow \begin{pmatrix} tI_l & -X^T \\ -X & tI_k \end{pmatrix} \succeq 0.$$

Indeed, the result in question follows from the fact that the sums of p largest eigenvalues of a symmetric matrix are SDr (Example 18c) due to the following

Observation. *The singular values $\sigma_i(X)$ of a rectangular $k \times l$ matrix X ($k \leq l$) for $i \leq k$ are equal to the eigenvalues $\lambda_i(\bar{X})$ of the $(k+l) \times (k+l)$ symmetric matrix*

$$\bar{X} = \begin{pmatrix} 0 & X^T \\ X & 0 \end{pmatrix}.$$

Since \bar{X} linearly depends on X , SDR's of the functions $S_p(\cdot)$ induce SDR's of the functions $\Sigma_p(X) = S_p(\bar{X})$ (Rule on affine substitution, Lecture 2; recall that all “calculus rules” established in Lecture 2 for CQR's are valid for SDR's as well).

Let us justify our observation. Let $X = \sum_{i=1}^k \sigma_i(X) e_i f_i^T$ be a singular value decomposition of X . We claim that the $2k$ $(k+l)$ -dimensional vectors $g_i^+ = \begin{pmatrix} f_i \\ e_i \end{pmatrix}$ and $g_i^- = \begin{pmatrix} f_i \\ -e_i \end{pmatrix}$ are orthogonal to each other, and they are eigenvectors of \bar{X} with the eigenvalues $\sigma_i(X)$ and $-\sigma_i(X)$, respectively. Moreover, \bar{X} vanishes on the orthogonal complement of the linear span of these vectors. In other words, we claim that the eigenvalues of \bar{X} , arranged in the non-ascending order, are as follows:

$$\sigma_1(X), \sigma_2(X), \dots, \sigma_k(X), \underbrace{0, \dots, 0}_{2(l-k)}, -\sigma_k(X), -\sigma_{k-1}(X), \dots, -\sigma_1(X);$$

this, of course, proves our Observation.

Now, the fact that the $2k$ vectors g_i^\pm , $i = 1, \dots, k$, are mutually orthogonal and nonzero is evident. Furthermore (we write σ_i instead of $\sigma_i(X)$),

$$\begin{aligned} \begin{pmatrix} 0 & X^T \\ X & 0 \end{pmatrix} \begin{pmatrix} f_i \\ e_i \end{pmatrix} &= \begin{pmatrix} 0 & \sum_{j=1}^k \sigma_j f_j e_j^T \\ \sum_{j=1}^k \sigma_j e_j f_j^T & 0 \end{pmatrix} \begin{pmatrix} f_i \\ e_i \end{pmatrix} \\ &= \begin{pmatrix} \sum_{j=1}^k \sigma_j f_j (e_j^T e_i) \\ \sum_{j=1}^k \sigma_j e_j (f_j^T f_i) \end{pmatrix} \\ &= \sigma_i \begin{pmatrix} f_i \\ e_i \end{pmatrix} \end{aligned}$$

(we have used that both $\{f_j\}$ and $\{e_j\}$ are orthonormal systems). Thus, g_i^+ is an eigenvector of \bar{X} with the eigenvalue $\sigma_i(X)$. Similar computation shows that g_i^- is an eigenvector of \bar{X} with the eigenvalue $-\sigma_i(X)$.

It remains to verify that if $h = \begin{pmatrix} f \\ e \end{pmatrix}$ is orthogonal to all g_i^\pm (f is l -dimensional, e is k -dimensional), then $\bar{X}h = 0$. Indeed, the orthogonality assumption means that $f^T f_i \pm e^T e_i = 0$ for all i , whence $e^T e_i = 0$ and $f^T f_i = 0$ for all i . Consequently,

$$\begin{pmatrix} 0 & X^T \\ X & 0 \end{pmatrix} \begin{pmatrix} f \\ e \end{pmatrix} = \begin{pmatrix} \sum_{i=1}^k f_i (e_i^T e) \\ \sum_{i=1}^k e_i (f_i^T f) \end{pmatrix} = 0. \quad \blacksquare$$

Looking at Proposition 3.2.1, we see that the fact that specific functions of eigenvalues of a symmetric matrix X , namely, the sums $S_k(X)$ of k largest eigenvalues of X , are SDr, underlies the possibility to build SDR's for a wide class of functions of the eigenvalues. The role of the sums of k largest singular values of a rectangular matrix X is equally important:

Proposition 3.2.2 *Let $g(x_1, \dots, x_k) : \mathbf{R}_+^k \rightarrow \mathbf{R} \cup \{+\infty\}$ be a symmetric monotone function:*

$$0 \leq y \leq x \in \text{Dom } f \Rightarrow f(y) \leq f(x).$$

Assume that g is SDr:

$$t \geq g(x) \Leftrightarrow \exists u : \mathcal{S}(x, t, u) \succeq 0,$$

with \mathcal{S} affinely depending on x, t, u . Then the function

$$f(X) = g(\sigma(X))$$

of $k \times l$ ($k \leq l$) rectangular matrix X is SDr, with SDR given by the relation

$$\begin{aligned} (a) \quad & t \geq f(X) \\ & \Updownarrow \\ & \exists x_1, \dots, x_k, u : \\ (b) \quad & \left\{ \begin{array}{l} \mathcal{S}(x_1, \dots, x_k, t, u) \succeq 0 \\ x_1 \geq x_2 \geq \dots \geq x_k \\ \Sigma_j(X) \leq x_1 + \dots + x_j, \quad j = 1, \dots, m \end{array} \right. \end{aligned} \tag{3.2.4}$$

Note the difference between the symmetric (Proposition 3.2.1) and the non-symmetric (Proposition 3.2.2) situations: in the former the function $g(x)$ was assumed to be SDr and symmetric only, while in the latter the monotonicity requirement is added.

The proof of Proposition 3.2.2 is outlined in Section 3.8

“Nonlinear matrix inequalities”. There are several cases when matrix inequalities $F(x) \succeq 0$, where F is a nonlinear function of x taking values in the space of symmetric $m \times m$ matrices, can be “linearized” – expressed via LMI's.

20a. General quadratic matrix inequality. Let X be a rectangular $k \times l$ matrix and

$$F(X) = (AXB)(AXB)^T + CXD + (CXD)^T + E$$

be a “quadratic” matrix-valued function of X ; here $A, B, C, D, E = E^T$ are rectangular matrices of appropriate sizes. Let m be the row size of the values of F . Consider the “ \succeq -epigraph” of the (matrix-valued!) function F – the set

$$\{(X, Y) \in \mathbf{M}^{k,l} \times \mathbf{S}^m \mid F(X) \preceq Y\}.$$

We claim that this set is SDr with the SDR

$$\left(\begin{array}{c|c} I_r & (AXB)^T \\ \hline AXB & Y - E - CXD - (CXD)^T \end{array} \right) \succeq 0 \quad [B : l \times r]$$

Indeed, by the Schur Complement Lemma our LMI is satisfied if and only if the Schur complement of the North-Western block is positive semidefinite, which is exactly our original “quadratic” matrix inequality.

20b. General “fractional-quadratic” matrix inequality. Let X be a rectangular $k \times l$ matrix, and V be a positive definite symmetric $l \times l$ matrix. Then we can define the matrix-valued function

$$F(X, V) = XV^{-1}X^T$$

taking values in the space of $k \times k$ symmetric matrices. We claim that the closure of the \succeq -epigraph of this (matrix-valued!) function, i.e., the set

$$E = \text{cl } \{(X, V; Y) \in \mathbf{M}^{k,l} \times \mathbf{S}_{++}^l \times \mathbf{S}^k \mid F(X, V) \equiv XV^{-1}X^T \preceq Y\}$$

is SDr, and an SDR of this set is given by the LMI

$$\left(\begin{array}{cc} V & X^T \\ X & Y \end{array} \right) \succeq 0. \quad (\text{R})$$

Indeed, by the Schur Complement Lemma a triple (X, V, Y) with *positive definite* V belongs to the “epigraph of F ” – satisfies the relation $F(X, V) \preceq Y$ – if and only if it satisfies (R). Now, if a triple (X, V, Y) belongs to E , i.e., it is the limit of a sequence of triples from the epigraph of F , then it satisfies (R) (as a limit of triples satisfying (R)). Vice versa, if a triple (X, V, Y) satisfies (R), then V is positive semidefinite (as a diagonal block in a positive semidefinite matrix). The “regularized” triples $(X, V_\epsilon = V + \epsilon I_l, Y)$ associated with $\epsilon > 0$ satisfy (R) along with the triple (X, V, R) ; since, as we just have seen, $V \succeq 0$, we have $V_\epsilon \succ 0$, for $\epsilon > 0$. Consequently, the triples (X, V_ϵ, Y) belong to E (this was our very first observation); since the triple (X, V, Y) is the limit of the regularized triples which, as we have seen, all belong to the epigraph of F , the triple (X, Y, V) belongs to the closure E of this epigraph. ■

20c. Matrix inequality $Y \preceq (C^T X^{-1} C)^{-1}$. In the case of scalars x, y the inequality $y \leq (cx^{-1}c)^{-1}$ in variables x, y is just an awkward way to write down the linear inequality $y \leq c^{-2}x$, but it naturally to the matrix analogy of the original inequality, namely, $Y \preceq (C^T X^{-1} C)^{-1}$, with rectangular $m \times n$ matrix C and variable symmetric $n \times n$ matrix Y and $m \times m$ matrix X . In order for the matrix inequality to make sense, we should assume that the rank of C equals n (and thus $m \geq n$). Under this assumption, the matrix $(C^T X^{-1} C)^{-1}$ makes sense at least for a positive definite X . We claim that the closure of the solution set of the resulting inequality – the set

$$\mathcal{X} = \text{cl} \{(X, Y) \in \mathbf{S}^m \times \mathbf{S}^n \mid X \succ 0, Y \preceq (C^T X^{-1} C)^{-1}\}$$

is SDR:

$$\mathcal{X} = \{(X, Y) \mid \exists Z : Y \preceq Z, Z \succeq 0, X \succeq CZC^T\}.$$

Indeed, let us denote by \mathcal{X}' the set in the right hand side of the latter relation; we should prove that $\mathcal{X}' = \mathcal{X}$. By definition, \mathcal{X} is the closure of its intersection with the domain $X \succ 0$. It is clear that \mathcal{X}' also is the closure of its intersection with the domain $X \succ 0$. Thus, all we need to prove is that a pair (Y, X) with $X \succ 0$ belongs to \mathcal{X} if and only if it belongs to \mathcal{X}' .

“If” part: Assume that $X \succ 0$ and $(Y, X) \in \mathcal{X}'$. Then there exists Z such that $Z \succeq 0$, $Z \succeq Y$ and $X \succeq CZC^T$. Let us choose a sequence $Z_i \succ Z$ such that $Z_i \rightarrow Z$, $i \rightarrow \infty$. Since $CZ_i C^T \rightarrow CZC^T \preceq X$ as $i \rightarrow \infty$, we can find a sequence of matrices X_i such that $X_i \rightarrow X$, $i \rightarrow \infty$, and $X_i \succ CZ_i C^T$ for all i . By the Schur Complement Lemma, the matrices $\begin{pmatrix} X_i & C \\ C^T & Z_i^{-1} \end{pmatrix}$ are positive definite; applying this lemma again, we conclude that $Z_i^{-1} \succeq C^T X_i^{-1} C$. Note that the left and the right hand side matrices in the latter inequality are positive definite. Now let us use the following simple fact

Lemma 3.2.2 *Let U, V be positive definite matrices of the same size. Then*

$$U \preceq V \Leftrightarrow U^{-1} \succeq V^{-1}.$$

Proof. Note that we can multiply an inequality $A \preceq B$ by a matrix Q from the left and Q^T from the right:

$$A \preceq B \Rightarrow QAQ^T \preceq QBQ^T \quad [A, B \in \mathbf{S}^m, Q \in \mathbf{M}^{k,m}]$$

(why?) Thus, if $0 \prec U \preceq V$, then $V^{-1/2}UV^{-1/2} \preceq V^{-1/2}VV^{-1/2} = I$ (note that $V^{-1/2} = [V^{-1/2}]^T$), whence clearly $V^{1/2}U^{-1}V^{1/2} = [V^{-1/2}UV^{-1/2}]^{-1} \succeq I$. Thus, $V^{1/2}U^{-1}V^{1/2} \succeq I$; multiplying this inequality from the left and from the right by $V^{-1/2} = [V^{-1/2}]^T$, we get $U^{-1} \succeq V^{-1}$. ■

Applying Lemma 3.2.2 to the inequality $Z_i^{-1} \succeq C^T X_i^{-1} C [\succ 0]$, we get $Z_i \preceq (C^T X_i^{-1} C)^{-1}$. As $i \rightarrow \infty$, the left hand side in this inequality converges to Z , and the right hand side converges to $(C^T X^{-1} C)^{-1}$. Hence $Z \preceq (C^T X^{-1} C)^{-1}$, and since $Y \preceq Z$, we get $Y \preceq (C^T X^{-1} C)^{-1}$, as claimed.

“Only if” part: Let $X \succ 0$ and $Y \preceq (C^T X^{-1} C)^{-1}$; we should prove that there exists $Z \succeq 0$ such that $Z \succeq Y$ and $X \succeq CZC^T$. We claim that the required relations are satisfied by $Z = (C^T X^{-1} C)^{-1}$. The only nontrivial part of the claim is that $X \succeq CZC^T$, and here is the required justification: by its origin $Z \succ 0$, and by the Schur Complement Lemma the matrix $\begin{pmatrix} Z^{-1} & C^T \\ C & X \end{pmatrix}$ is positive semidefinite, whence, by the same Lemma, $X \succeq C(Z^{-1})^{-1}C^T = CZC^T$.

Nonnegative polynomials. Consider the problem of the best polynomial approximation – given a function f on certain interval, we want to find its best uniform (or Least Squares, etc.) approximation by a polynomial of a given degree. This problem arises typically as a subproblem in all kinds of signal processing problems. In some situations the approximating polynomial is required to be nonnegative (think, e.g., of the case where the resulting polynomial is an estimate of an unknown probability density); how to express the nonnegativity restriction? As it was shown by Yu. Nesterov [26], it can be done via semidefinite programming:

The set of all nonnegative (on the entire axis, or on a given ray, or on a given segment) polynomials of a given degree is SDr.

In this statement (and everywhere below) we identify a polynomial $p(t) = \sum_{i=0}^k p_i t^i$ of degree (not exceeding) k with the $(k+1)$ -dimensional vector $\text{Coef}(p) = \text{Coef}(p) = (p_0, p_1, \dots, p_k)^T$ of the coefficients of p . Consequently, a set of polynomials of the degree $\leq k$ becomes a set in \mathbf{R}^{k+1} , and we may ask whether this set is or is not SDr.

Let us look what are the SDR's of different sets of nonnegative polynomials. The key here is to get a SDR for the set $P_{2k}^+(\mathbf{R})$ of polynomials of (at most) a given degree $2k$ which are nonnegative on the entire axis⁶⁾

21a. Polynomials nonnegative on the entire axis: The set $P_{2k}^+(\mathbf{R})$ is SDr – it is the image of the semidefinite cone \mathbf{S}_+^{k+1} under the affine mapping

$$X \mapsto \text{Coef}(e^T(t) X e(t)) : \mathbf{S}^{k+1} \rightarrow \mathbf{R}^{2k+1}, \quad e(t) = \begin{pmatrix} 1 \\ t \\ t^2 \\ \vdots \\ t^k \end{pmatrix} \quad (\text{C})$$

First note that the fact that $P^+ \equiv P_{2k}^+(\mathbf{R})$ is an affine image of the semidefinite cone indeed implies the SD-representability of P^+ , see the “calculus” of conic representations in Lecture 2. Thus, all we need is to show that P^+ is exactly the same as the image, let it be called P , of \mathbf{S}_+^{k+1} under the mapping (C).

(1) The fact that P is contained in P^+ is immediate. Indeed, let X be a $(k+1) \times (k+1)$ positive semidefinite matrix. Then X is a sum of dyadic matrices:

$$X = \sum_{i=1}^{k+1} p^i (p^i)^T, \quad p^i = (p_{i0}, p_{i1}, \dots, p_{ik})^T \in \mathbf{R}^{k+1}$$

(why?) But then

$$e^T(t) X e(t) = \sum_{i=1}^{k+1} e^T(t) p^i [p^i]^T e(t) = \sum_{i=1}^{k+1} \left(\sum_{j=0}^k p_{ij} t^j \right)^2$$

is the sum of squares of other polynomials and therefore is nonnegative on the axis. Thus, the image of X under the mapping (C) belongs to P^+ .

Note that reversing our reasoning, we get the following result:

⁶⁾ It is clear why we have restricted the degree to be even: a polynomial of an odd degree cannot be nonnegative on the entire axis!

(!) If a polynomial $p(t)$ of degree $\leq 2k$ can be represented as a sum of squares of other polynomials, then the vector $\text{Coef}(p)$ of the coefficients of p belongs to the image of \mathbf{S}_+^{k+1} under the mapping (C).

With (!), the remaining part of the proof – the demonstration that the image of \mathbf{S}_+^{k+1} contains P^+ , is readily given by the following well-known algebraic fact:

(!!) A polynomial is nonnegative on the axis if and only if it is a sum of squares of polynomials.

The proof of (!!) is so nice that we cannot resist the temptation to present it here. The “if” part is evident. To prove the “only if” one, assume that $p(t)$ is nonnegative on the axis, and let the degree of p (it must be even) be $2k$. Now let us look at the roots of p . The real roots $\lambda_1, \dots, \lambda_r$ must be of even multiplicities $2m_1, 2m_2, \dots, 2m_r$ each (otherwise p would alter its sign in a neighbourhood of a root, which contradicts the nonnegativity). The complex roots of p can be arranged in conjugate pairs $(\mu_1, \mu_1^*), (\mu_2, \mu_2^*), \dots, (\mu_s, \mu_s^*)$, and the factor of p

$$(t - \mu_i)(t - \mu_i^*) = (t - \Re \mu_i)^2 + (\Im \mu_i)^2$$

corresponding to such a pair is a sum of two squares. Finally, the leading coefficient of p is positive. Consequently, we have

$$p(t) = \omega^2 [(t - \lambda_1)^2]^{m_1} \dots [(t - \lambda_r)^2]^{m_r} [(t - \mu_1)(t - \mu_1^*)] \dots [(t - \mu_s)(t - \mu_s^*)]$$

is a product of sums of squares. But such a product is itself a sum of squares (open the parentheses)!

In fact we can say more: a nonnegative polynomial p is a sum of just two squares! To see this, note that, as we have seen, p is a product of sums of two squares and take into account the following fact (Louville):

The product of sums of two squares is again a sum of two squares:

$$(a^2 + b^2)(c^2 + d^2) = (ac - bd)^2 + (ad + bc)^2$$

(cf. with: “the modulus of a product of two complex numbers is the product of their modulae”).

Equipped with the SDR of the set $P_{2k}^+(\mathbf{R})$ of polynomials nonnegative on the entire axis, we can immediately obtain SDR’s for the polynomials nonnegative on a given ray/segment:

21b. Polynomials nonnegative on a ray/segment.

1) The set $P_k^+(\mathbf{R}_+)$ of (coefficients of) polynomials of degree $\leq k$ which are nonnegative on the nonnegative ray, is SDr.

Indeed, this set is the inverse image of the SDr set $P_{2k}^+(\mathbf{R})$ under the linear mapping of the spaces of (coefficients of) polynomials given by the mapping

$$p(t) \mapsto p^+(t) \equiv p(t^2)$$

(recall that the inverse image of an SDr set is SDr).

2) The set $P_k^+([0, 1])$ of (coefficients of) polynomials of degree $\leq k$ which are nonnegative on the segment $[0, 1]$, is SDr.

Indeed, a polynomial $p(t)$ of degree $\leq k$ is nonnegative on $[0, 1]$ if and only if the rational function

$$g(t) = p\left(\frac{t^2}{1+t^2}\right)$$

is nonnegative on the entire axis, or, which is the same, if and only if the polynomial

$$p^+(t) = (1 + t^2)^k g(t)$$

of degree $\leq 2k$ is nonnegative on the entire axis. The coefficients of p^+ depend linearly on the coefficients of p , and we conclude that $P_k^+([0, 1])$ is the inverse image of the SDr set $P_{2k}^+(\mathbf{R})$ under certain linear mapping.

Our last example in this series deals with trigonometric polynomials

$$p(\phi) = a_0 + \sum_{\ell=1}^k [a_\ell \cos(\ell\phi) + b_\ell \sin(\ell\phi)]$$

Identifying such a polynomial with its vector of coefficients $\text{Coef}(p) \in \mathbf{R}^{2k+1}$, we may ask how to express the set $S_k^+(\Delta)$ of those trigonometric polynomials of degree $\leq k$ which are nonnegative on a segment $\Delta \subset [0, 2\pi]$.

21c. Trigonometric polynomials nonnegative on a segment. The set $S_k^+(\Delta)$ is SDr.

Indeed, $\sin(\ell\phi)$ and $\cos(\ell\phi)$ are polynomials of $\sin(\phi)$ and $\cos(\phi)$, and the latter functions, in turn, are rational functions of $\zeta = \tan(\phi/2)$:

$$\cos(\phi) = \frac{1 - \zeta^2}{1 + \zeta^2}, \sin(\phi) = \frac{2\zeta}{1 + \zeta^2} \quad [\zeta = \tan(\phi/2)].$$

Consequently, a trigonometric polynomial $p(\phi)$ of degree $\leq k$ can be represented as a rational function of $\zeta = \tan(\phi/2)$:

$$p(\phi) = \frac{p^+(\zeta)}{(1 + \zeta^2)^k} \quad [\zeta = \tan(\phi/2)],$$

where the coefficients of the algebraic polynomial p^+ of degree $\leq 2k$ are linear functions of the coefficients of p . Now, the requirement for p to be nonnegative on a given segment $\Delta \subset [0, 2\pi]$ is equivalent to the requirement for p^+ to be nonnegative on a “segment” Δ^+ (which, depending on Δ , may be either the usual finite segment, or a ray, or the entire axis). We see that $S_k^+(\Delta)$ is inverse image, under certain linear mapping, of the SDr set $P_{2k}^+(\Delta^+)$, so that $S_k^+(\Delta)$ itself is SDr.

Finally, we may ask which part of the above results can be saved when we pass from nonnegative polynomials of one variable to those of two or more variables. Unfortunately, not too much. E.g., among nonnegative polynomials of a given degree with $r > 1$ variables, exactly those of them who are sums of squares can be obtained as the image of a positive semidefinite cone under certain linear mapping similar to (D). The difficulty is that in the multi-dimensional case the nonnegativity of a polynomial is not equivalent to its representability as a sum of squares, thus, the positive semidefinite cone gives only part of the polynomials we are interested to describe.

3.3 Applications of Semidefinite Programming in Engineering

Due to its tremendous expressive abilities, Semidefinite Programming allows to pose and process numerous highly nonlinear convex optimization programs arising in applications, in particular, in Engineering. We are about to outline briefly just few instructive examples.

3.3.1 Dynamic Stability in Mechanics

“Free motions” of the so called *linearly elastic* mechanical systems, i.e., their behaviour when no external forces are applied, are governed by systems of differential equations of the type

$$M \frac{d^2}{dt^2} x(t) = -Ax(t), \quad (\text{N})$$

where $x(t) \in \mathbf{R}^n$ is the state vector of the system at time t , M is the (generalized) “mass matrix”, and A is the “stiffness” matrix of the system. Basically, (N) is the Newton law for a system with the potential energy $\frac{1}{2}x^T Ax$.

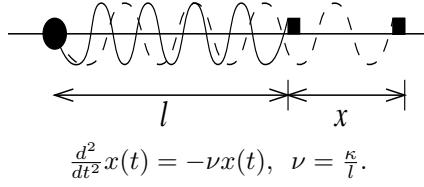
As a simple example, consider a system of k points of masses μ_1, \dots, μ_k linked by springs with given elasticity coefficients; here x is the vector of the displacements $x_i \in \mathbf{R}^d$ of the points from their equilibrium positions e_i ($d = 1/2/3$ is the dimension of the model). The Newton equations become

$$\mu_i \frac{d^2}{dt^2} x_i(t) = - \sum_{j \neq i} \nu_{ij} (e_i - e_j)(e_i - e_j)^T (x_i - x_j), \quad i = 1, \dots, k,$$

with ν_{ij} given by

$$\nu_{ij} = \frac{\kappa_{ij}}{\|e_i - e_j\|_2^3},$$

where $\kappa_{ij} > 0$ are the elasticity coefficients of the springs. The resulting system is of the form (N) with a diagonal matrix M and a positive semidefinite symmetric matrix A . The well-known simplest system of this type is a *pendulum* (a single point capable to slide along a given axis and linked by a spring to a fixed point on the axis):



Another example is given by *trusses* – mechanical constructions, like a railway bridge or the Eiffel Tower, built from linked to each other thin elastic bars.

Note that in the above examples both the mass matrix M and the stiffness matrix A are symmetric positive semidefinite; in “nondegenerate” cases they are even positive definite, and this is what we assume from now on. Under this assumption, we can pass in (N) from the variables $x(t)$ to the variables $y(t) = M^{1/2}x(t)$; in these variables the system becomes

$$\frac{d^2}{dt^2} y(t) = -\hat{A}y(t), \quad \hat{A} = M^{-1/2}AM^{-1/2}. \quad (\text{N}')$$

It is well known that the space of solutions of system (N') (where \hat{A} is symmetric positive definite) is spanned by fundamental (perhaps complex-valued) solutions of the form $\exp\{\mu t\}f$. A nontrivial (with $f \neq 0$) function of this type is a solution to (N') if and only if

$$(\mu^2 I + \hat{A})f = 0,$$

so that the allowed values of μ^2 are the minus eigenvalues of the matrix \hat{A} , and f 's are the corresponding eigenvectors of \hat{A} . Since the matrix \hat{A} is symmetric positive definite, the only

allowed values of μ are purely imaginary, with the imaginary parts $\pm\sqrt{\lambda_j(\hat{A})}$. Recalling that the eigenvalues/eigenvectors of \hat{A} are exactly the eigenvalues/eigenvectors of the pencil $[M, A]$, we come to the following result:

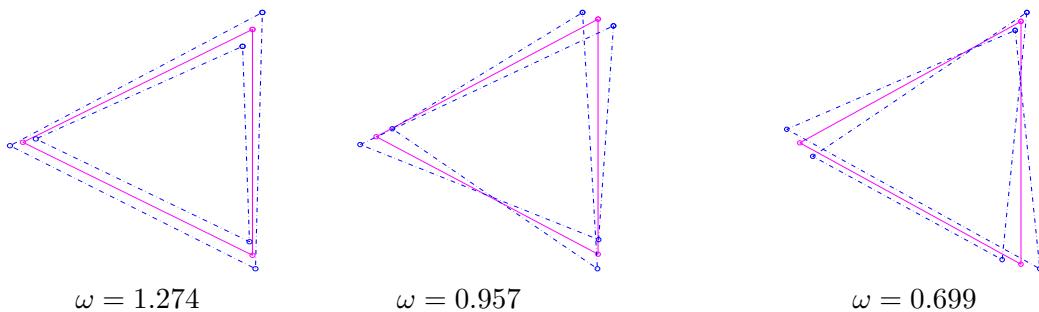
(!) In the case of positive definite symmetric M, A , the solutions to (N) – the “free motions” of the corresponding mechanical system \mathcal{S} – are of the form

$$x(t) = \sum_{j=1}^n [a_j \cos(\omega_j t) + b_j \sin(\omega_j t)] e_j,$$

where a_j, b_j are free real parameters, e_j are the eigenvectors of the pencil $[M, A]$:

$$(\lambda_j M - A)e_j = 0$$

and $\omega_j = \sqrt{\lambda_j}$. Thus, the “free motions” of the system \mathcal{S} are mixtures of harmonic oscillations along the eigenvectors of the pencil $[M, A]$, and the frequencies of the oscillations (“the eigenfrequencies of the system”) are the square roots of the corresponding eigenvalues of the pencil.



“Nontrivial” modes of a spring triangle (3 unit masses linked by springs)

Shown are 3 “eigenmotions” (modes) of a spring triangle with nonzero frequencies; at each picture, the dashed lines depict two instant positions of the oscillating triangle.

There are 3 more “eigenmotions” with zero frequency, corresponding to shifts and rotation of the triangle.

From the engineering viewpoint, the “dynamic behaviour” of mechanical constructions such as buildings, electricity masts, bridges, etc., is the better the larger are the eigenfrequencies of the system⁷⁾. This is why a typical design requirement in mechanical engineering is a lower bound

$$\lambda_{\min}(A : M) \geq \lambda_* \quad [\lambda_* > 0] \quad (3.3.1)$$

on the *smallest* eigenvalue $\lambda_{\min}(A : M)$ of the pencil $[M, A]$ comprised of the mass and the stiffness matrices of the would-be system. In the case of positive definite symmetric mass matrices (3.3.1) is equivalent to the matrix inequality

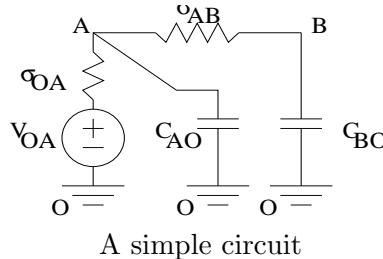
$$A - \lambda_* M \succeq 0. \quad (3.3.2)$$

⁷⁾Think about a building and an earthquake or about sea waves and a light house: in this case the external load acting at the system is time-varying and can be represented as a sum of harmonic oscillations of different (and low) frequencies; if some of these frequencies are close to the eigenfrequencies of the system, the system can be crushed by resonance. In order to avoid this risk, one is interested to move the eigenfrequencies of the system away from 0 as far as possible.

If M and A are affine functions of the design variables (as is the case in, e.g., Truss Design), the matrix inequality (3.3.2) is a *linear* matrix inequality on the design variables, and therefore it can be processed via the machinery of semidefinite programming. Moreover, in the cases when A is affine in the design variables, and M is constant, (3.3.2) is an LMI in the design variables and λ_* , and we may play with λ_* , e.g., solve a problem of the type “given the mass matrix of the system to be designed and a number of (SDr) constraints on the design variables, build a system with the minimum eigenfrequency as large as possible”, which is a semidefinite program, provided that the stiffness matrix is affine in the design variables.

3.3.2 Design of chips and Boyd's time constant

Consider an RC-electric circuit, i.e., a circuit comprised of three types of elements: (1) resistors; (2) capacitors; (3) resistors in a series combination with outer sources of voltage:



A simple circuit

Element OA: outer supply of voltage V_{OA} and resistor with conductance σ_{OA}

Element AO: capacitor with capacitance C_{AO}

Element AB: resistor with conductance σ_{AB}

Element BO: capacitor with capacitance C_{BO}

E.g., a chip is, electrically, a complicated circuit comprised of elements of the indicated type. When designing chips, the following characteristics are of primary importance:

- *Speed.* In a chip, the outer voltages are switching at certain frequency from one constant value to another. Every switch is accompanied by a “transition period”; during this period, the potentials/currents in the elements are moving from their previous values (corresponding to the static steady state for the “old” outer voltages) to the values corresponding to the new static steady state. Since there are elements with “inertia” – capacitors – this transition period takes some time⁸⁾. In order to ensure stable performance of the chip, the transition period should be much less than the time between subsequent switches in the outer voltages. Thus, the duration of the transition period is responsible for the speed at which the chip can perform.
- *Dissipated heat.* Resistors in the chip dissipate heat which should be eliminated, otherwise the chip will not function. This requirement is very serious for modern “high-density” chips. Thus, a characteristic of vital importance is the dissipated heat power.

The two objectives: high speed (i.e., a small transition period) and small dissipated heat – usually are conflicting. As a result, a chip designer faces the tradeoff problem like “how to get a chip with a given speed and with the minimal dissipated heat”. It turns out that different

⁸⁾From purely mathematical viewpoint, the transition period takes infinite time – the currents/voltages approach asymptotically the new steady state, but never actually reach it. From the engineering viewpoint, however, we may think that the transition period is over when the currents/voltages become close enough to the new static steady state.

optimization problems related to the tradeoff between the speed and the dissipated heat in an RC circuit belong to the “semidefinite universe”. We restrict ourselves with building an SDR for the speed.

Simple considerations, based on Kirchoff laws, demonstrate that the transition period in an RC circuit is governed by a linear system of differential equations as follows:

$$C \frac{d}{dt} w(t) = -Sw(t) + Rv. \quad (3.3.3)$$

Here

- The state vector $w(\cdot)$ is comprised of the potentials at all but one nodes of the circuit (the potential at the remaining node – “the ground” – is normalized to be identically zero);
- Matrix $C \succeq 0$ is readily given by the topology of the circuit and the capacitances of the capacitors and is linear in these capacitances. Similarly, matrix $S \succeq 0$ is readily given by the topology of the circuit and the conductances of the resistors and is linear in these conductances. Matrix R is given solely by the topology of the circuit;
- v is the vector of outer voltages; recall that this vector is set to certain constant value at the beginning of the transition period.

As we have already mentioned, the matrices C and S , due to their origin, are positive semidefinite; in nondegenerate cases, they are even positive definite, which we assume from now on.

Let \hat{w} be the steady state of (3.3.3), so that $S\hat{w} = Rv$. The difference $\delta(t) = w(t) - \hat{w}$ is a solution to the homogeneous differential equation

$$C \frac{d}{dt} \delta(t) = -S\delta(t). \quad (3.3.4)$$

Setting $\gamma(t) = C^{1/2}\delta(t)$ (cf. Section 3.3.1), we get

$$\frac{d}{dt} \gamma(t) = -(C^{-1/2}SC^{-1/2})\gamma(t). \quad (3.3.5)$$

Since C and S are positive definite, all eigenvalues λ_i of the symmetric matrix $C^{-1/2}SC^{-1/2}$ are positive. It is clear that the space of solutions to (3.3.5) is spanned by the “eigenmotions”

$$\gamma_i(t) = \exp\{-\lambda_i t\} e_i,$$

where $\{e_i\}$ is an orthonormal eigenbasis of the matrix $C^{-1/2}SC^{-1/2}$. We see that all solutions to (3.3.5) (and thus - to (3.3.4) as well) are exponentially fast converging to 0, or, which is the same, the state $w(t)$ of the circuit exponentially fast approaches the steady state \hat{w} . The “time scale” of this transition is, essentially, defined by the quantity $\lambda_{\min} = \min_i \lambda_i$; a typical “decay rate” of a solution to (3.3.5) is nothing but $T = \lambda_{\min}^{-1}$. S. Boyd has proposed to use T to quantify the length of the transition period, and to use the reciprocal of it – i.e., the quantity λ_{\min} itself – as the quantitative measure of the speed. Technically, the main advantage of this definition is that the speed turns out to be the minimum eigenvalue of the matrix $C^{-1/2}SC^{-1/2}$, i.e., the minimum eigenvalue of the matrix pencil $[C : S]$. Thus, the speed in Boyd’s definition turns out to be efficiently computable (which is not the case for other, more sophisticated, “time constants” used by engineers). Even more important, with Boyd’s approach, a typical design

specification “the speed of a circuit should be at least such and such” is modelled by the matrix inequality

$$S \succeq \lambda_* C. \quad (3.3.6)$$

As it was already mentioned, S and C are linear in the capacitances of the capacitors and conductances of the resistors; in typical circuit design problems, the latter quantities are affine functions of the design parameters, and (3.3.6) becomes an LMI in the design parameters.

3.3.3 Lyapunov stability analysis/synthesis

3.3.3.1 Uncertain dynamical systems

Consider a time-varying *uncertain* linear dynamical system

$$\frac{d}{dt}x(t) = A(t)x(t), \quad x(0) = x_0. \quad (\text{ULS})$$

Here $x(t) \in \mathbf{R}^n$ represents the state of the system at time t , and $A(t)$ is a time-varying $n \times n$ matrix. We assume that the system is *uncertain* in the sense that we have no idea of what is x_0 , and all we know about $A(t)$ is that this matrix, at any time t , belongs to a given uncertainty set \mathcal{U} . Thus, (ULS) represents a wide family of linear dynamic systems rather than a single system; and it makes sense to call a *trajectory* of the uncertain linear system (ULS) every function $x(t)$ which is an “actual trajectory” of a system from the family, i.e., is such that

$$\frac{d}{dt}x(t) = A(t)x(t)$$

for all $t \geq 0$ and certain matrix-valued function $A(t)$ taking all its values in \mathcal{U} .

Note that we can model a *nonlinear* dynamic system

$$\frac{d}{dt}x(t) = f(t, x(t)) \quad [x \in \mathbf{R}^n] \quad (\text{NLS})$$

with a given right hand side $f(t, x)$ and a given equilibrium $x(t) \equiv 0$ (i.e., $f(t, 0) = 0$, $t \geq 0$) as an uncertain *linear* system. Indeed, let us define the set \mathcal{U}_f as the closed convex hull of the set of $n \times n$ matrices $\{\frac{\partial}{\partial x}f(t, x) \mid t \geq 0, x \in \mathbf{R}^n\}$. Then for every point $x \in \mathbf{R}^n$ we have

$$\begin{aligned} f(t, x) &= f(t, 0) + \int_0^s \left[\frac{\partial}{\partial x}f(t, sx) \right] x ds = A_x(t)x, \\ A_x(t) &= \int_0^1 \frac{\partial}{\partial x}f(t, sx) ds \in \mathcal{U}. \end{aligned}$$

We see that every trajectory of the original nonlinear system (NLS) is also a trajectory of the uncertain linear system (ULS) associated with the uncertainty set $\mathcal{U} = \mathcal{U}_f$ (this trick is called “global linearization”). Of course, the set of trajectories of the resulting uncertain linear system can be much wider than the set of trajectories of (NLS); however, all “good news” about the uncertain system (like “all trajectories of (ULS) share such and such property”) are automatically valid for the trajectories of the “nonlinear system of interest” (NLS), and only “bad news” about (ULS) (“such and such property is not shared by some trajectories of (ULS)”) may say nothing about the system of interest (NLS).

3.3.3.2 Stability and stability certificates

The basic question about a dynamic system is the one of its stability. For (ULS), this question sounds as follows:

(?) Is it true that (ULS) is stable, i.e., that

$$x(t) \rightarrow 0 \text{ as } t \rightarrow \infty$$

for every trajectory of the system?

A sufficient condition for the stability of (ULS) is the existence of a quadratic Lyapunov function, i.e., a quadratic form $\mathcal{L}(x) = x^T X x$ with symmetric positive definite matrix X such that

$$\frac{d}{dt} \mathcal{L}(x(t)) \leq -\alpha \mathcal{L}(x(t)) \quad (3.3.7)$$

for certain $\alpha > 0$ and all trajectories of (ULS):

Lemma 3.3.1 [Quadratic Stability Certificate] Assume (ULS) admits a quadratic Lyapunov function \mathcal{L} . Then (ULS) is stable.

Proof. If (3.3.7) is valid with some $\alpha > 0$ for all trajectories of (ULS), then, by integrating this differential inequality, we get

$$\mathcal{L}(x(t)) \leq \exp\{-\alpha \mathcal{L}(x(0))\} \rightarrow 0 \text{ as } t \rightarrow \infty.$$

Since $\mathcal{L}(\cdot)$ is a positive definite quadratic form, $\mathcal{L}(x(t)) \rightarrow 0$ implies that $x(t) \rightarrow 0$. ■

Of course, the statement of Lemma 3.3.1 also holds for non-quadratic Lyapunov functions: all we need is (3.3.7) plus the assumption that $\mathcal{L}(x)$ is smooth, nonnegative and is bounded away from 0 outside every neighbourhood of the origin. The advantage of a quadratic Lyapunov function is that we more or less know how to find such a function, if it exists:

Proposition 3.3.1 [Existence of Quadratic Stability Certificate] Let \mathcal{U} be the uncertainty set associated with uncertain linear system (ULS). The system admits quadratic Lyapunov function if and only if the optimal value of the “semi-infinite”⁹⁾ semidefinite program

$$\begin{aligned} & \text{minimize} && s \\ & \text{s.t.} && \\ & && sI_n - A^T X - XA \succeq 0, \quad \forall A \in \mathcal{U} \\ & && X \succeq I_n \end{aligned} \quad (\text{Ly})$$

with the design variables $s \in \mathbf{R}$ and $X \in \mathbf{S}^n$, is negative. Moreover, every feasible solution to the problem with negative value of the objective provides a quadratic Lyapunov stability certificate for (ULS).

We shall refer to a positive definite matrix $X \succeq I_n$ which can be extended, by properly chosen $s < 0$, to a feasible solution of (Ly), as to a Lyapunov stability certificate for (ULS), the uncertainty set being \mathcal{U} .

Proof of Proposition 3.3.1. The derivative $\frac{d}{dt} [x^T(t) X x(t)]$ of the quadratic function $x^T X x$ along a trajectory of (ULS) is equal to

$$\left[\frac{d}{dt} x(t) \right]^T X x(t) + x^T(t) X \left[\frac{d}{dt} x(t) \right] = x^T(t) [A^T(t) X + X A(t)] x(t).$$

⁹⁾i.e., with infinitely many LMI constraints

If $x^T X x$ is a Lyapunov function, then the resulting quantity must be at most $-\alpha x^T(t) X x(t)$, i.e., we should have

$$x^T(t) \left[-\alpha X - A^T(t)X - XA(t) \right] x(t) \geq 0$$

for every possible value of $A(t)$ at any time t and for every possible value $x(t)$ of a trajectory of the system at this time. Since possible values of $x(t)$ fill the entire \mathbf{R}^n and possible values of $A(t)$ fill the entire \mathcal{U} , we conclude that

$$-\alpha X - A^T X - XA \succeq 0 \quad \forall A \in \mathcal{U}.$$

By definition of a quadratic Lyapunov function, $X \succ 0$ and $\alpha > 0$; by normalization (dividing both X and α by the smallest eigenvalue of X), we get a pair $\hat{s} > 0$, $\hat{X} \geq I_n$ such that

$$-\hat{s}\hat{X} - A^T\hat{X} - \hat{X}A \succeq 0 \quad \forall A \in \mathcal{U}.$$

Since $\hat{X} \succeq I_n$, we conclude that

$$-\hat{s}I_n - A^T\hat{X} - \hat{X}A \succeq \hat{s}\hat{X} - A^T\hat{X} - \hat{X}A \succeq 0 \quad \forall A \in \mathcal{U};$$

thus, $(s = -\hat{s}, \hat{X})$ is a feasible solution to (Ly) with negative value of the objective. We have demonstrated that if (ULS) admits a quadratic Lyapunov function, then (Ly) has a feasible solution with negative value of the objective. Reversing the reasoning, we can verify the inverse implication. ■

Lyapunov stability analysis. According to Proposition 3.3.1, the existence of a Lyapunov stability certificate is a *sufficient*, but, in general, not a necessary stability condition for (ULS). When the condition is not satisfied (i.e., if the optimal value in (Ly) is nonnegative), then all we can say is that *the stability of (ULS) cannot be certified by a quadratic Lyapunov function*, although (ULS) still may be stable.¹⁰⁾ In this sense, the stability analysis based on quadratic Lyapunov functions is conservative. This drawback, however, is in a sense compensated by the fact that this kind of stability analysis is “implementable”: in many cases we can efficiently solve (Ly), thus getting a quadratic “stability certificate”, provided that it exists, in a constructive way. Let us look at two such cases.

Polytopic uncertainty set. The first “tractable case” of (Ly) is when \mathcal{U} is a polytope given as a convex hull of finitely many points:

$$\mathcal{U} = \text{Conv}\{A_1, \dots, A_N\}.$$

In this case (Ly) is equivalent to the semidefinite program

$$\min_{s, X} \left\{ s : sI_n - A_i^T X - XA_i \succeq 0, \quad i = 1, \dots, N; X \succeq I_n \right\} \quad (3.3.8)$$

¹⁰⁾The only case when the existence of a quadratic Lyapunov function is a criterion (i.e., a *necessary and sufficient* condition) for stability is the simplest case of certain time-invariant linear system $\frac{d}{dt}x(t) = Ax(t)$ ($\mathcal{U} = \{A\}$). This is the case which led Lyapunov to the general concept of what is now called “a Lyapunov function” and what is the basic approach to establishing convergence of different time-dependent processes to their equilibria. Note also that in the case of time-invariant linear system there exists a straightforward algebraic stability criterion – all eigenvalues of A should have negative real parts. The advantage of the Lyapunov approach is that it can be extended to more general situations, which is not the case for the eigenvalue criterion.

(why?).

The assumption that \mathcal{U} is a polytope given as a convex hull of a finite set is crucial for a possibility to get a “computationally tractable” equivalent reformulation of (Ly). If \mathcal{U} is, say, a polytope given by a list of linear inequalities (e.g., all we know about the entries of $A(t)$ is that they reside in certain intervals; this case is called “interval uncertainty”), (Ly) may become as hard as a problem can be: it may happen that just to check whether a given pair (s, X) is feasible for (Ly) is already a “computationally intractable” problem. The same difficulties may occur when \mathcal{U} is a general-type ellipsoid in the space of $n \times n$ matrices. There exists, however, a specific type of “uncertainty ellipsoids” \mathcal{U} for which (Ly) is “easy”. Let us look at this case.

Norm-bounded perturbations. In numerous applications the $n \times n$ matrices A forming the uncertainty set \mathcal{U} are obtained from a fixed “nominal” matrix A_* by adding perturbations of the form $B\Delta C$, where $B \in \mathbf{M}^{n,k}$ and $C \in \mathbf{M}^{l,n}$ are given rectangular matrices and $\Delta \in \mathbf{M}^{k,l}$ is “the perturbation” varying in a “simple” set \mathcal{D} :

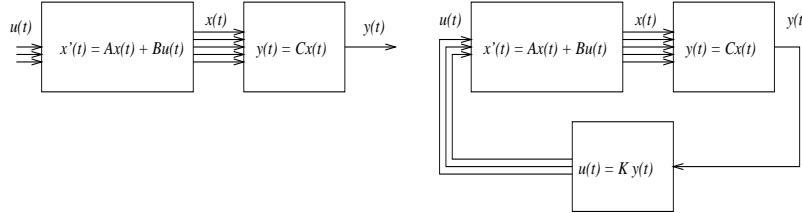
$$\mathcal{U} = \{A = A_* + B\Delta C \mid \Delta \in \mathcal{D} \subset \mathbf{M}^{k,l}\} \quad [B \in \mathbf{M}^{n,k}, 0 \neq C \in \mathbf{M}^{l,n}] \quad (3.3.9)$$

As an instructive example, consider a *controlled* linear time-invariant dynamic system

$$\begin{aligned} \frac{d}{dt}x(t) &= Ax(t) + Bu(t) \\ y(t) &= Cx(t) \end{aligned} \quad (3.3.10)$$

(x is the state, u is the control and y is the output we can observe) “closed” by a feedback

$$u(t) = Ky(t).$$



Open loop (left) and closed loop (right) controlled systems

The resulting “closed loop system” is given by

$$\frac{d}{dt}x(t) = \hat{A}x(t), \quad \hat{A} = A + BKC. \quad (3.3.11)$$

Now assume that A , B and C are constant and known, but the feedback K is drifting around certain nominal feedback K^* : $K = K^* + \Delta$. As a result, the matrix \hat{A} of the closed loop system also drifts around its nominal value $A^* = A + BK^*C$, and the perturbations in \hat{A} are exactly of the form $B\Delta C$.

Note that we could get essentially the same kind of drift in \hat{A} assuming, instead of additive perturbations, multiplicative perturbations $C = (I_l + \Delta)C^*$ in the observer (or multiplicative disturbances in the actuator B).

Now assume that the input perturbations Δ are of spectral norm $|\Delta|$ not exceeding a given ρ (*norm-bounded perturbations*):

$$\mathcal{D} = \{\Delta \in \mathbf{M}^{k,l} \mid |\Delta| \leq \rho\}. \quad (3.3.12)$$

Proposition 3.3.2 [8] *In the case of uncertainty set (3.3.9), (3.3.12) the “semi-infinite” semidefinite program (Ly) is equivalent to the usual semidefinite program*

$$\begin{array}{ll} \text{minimize} & \alpha \\ \text{s.t.} & \begin{pmatrix} \alpha I_n - A_*^T X - X A_* - \lambda C^T C & \rho X B \\ \rho B^T X & \lambda I_k \\ X & \succeq I_n \end{pmatrix} \succeq 0 \end{array} \quad (3.3.13)$$

in the design variables α, λ, X .

When shrinking the set of perturbations (3.3.12) to the ellipsoid

$$\mathcal{E} = \{\Delta \in \mathbf{M}^{k,l} \mid \|\Delta\|_2 \equiv \sqrt{\sum_{i=1}^k \sum_{j=1}^l \Delta_{ij}^2} \leq \rho\}, \quad ^{(11)} \quad (3.3.14)$$

we basically do not vary (Ly): in the case of the uncertainty set (3.3.9), (Ly) is still equivalent to (3.3.13).

Proof. It suffices to verify the following general statement:

Lemma 3.3.2 *Consider the matrix inequality*

$$Y - Q^T \Delta^T P^T Z^T R - R^T Z P \Delta Q \succeq 0 \quad (3.3.15)$$

where Y is symmetric $n \times n$ matrix, Δ is a $k \times l$ matrix and P, Q, Z, R are rectangular matrices of appropriate sizes (i.e., $q \times k$, $l \times n$, $p \times q$ and $p \times n$, respectively). Given Y, P, Q, Z, R , with $Q \neq 0$ (this is the only nontrivial case), this matrix inequality is satisfied for all Δ with $|\Delta| \leq \rho$ if and only if it is satisfied for all Δ with $\|\Delta\|_2 \leq \rho$, and this is the case if and only if

$$\begin{pmatrix} Y - \lambda Q^T Q & -\rho R^T Z P \\ -\rho P^T Z^T R & \lambda I_k \end{pmatrix} \succeq 0$$

for a properly chosen real λ .

The statement of Proposition 3.4.14 is just a particular case of Lemma 3.3.2. For example, in the case of uncertainty set (3.3.9), (3.3.12) a pair (α, X) is a feasible solution to (Ly) if and only if $X \succeq I_n$ and (3.3.15) is valid for $Y = \alpha X - A_*^T X - X A_*$, $P = B$, $Q = C$, $Z = X$, $R = I_n$; Lemma 3.3.2 provides us with an LMI reformulation of the latter property, and this LMI is exactly what we see in the statement of Proposition 3.4.14.

Proof of Lemma. (3.3.15) is valid for all Δ with $|\Delta| \leq \rho$ (let us call this property of (Y, P, Q, Z, R) “Property 1”) if and only if

$$2[\xi^T R^T Z P] \Delta [Q \xi] \leq \xi^T Y \xi \quad \forall \xi \in \mathbf{R}^n \quad \forall (\Delta : |\Delta| \leq \rho),$$

or, which is the same, if and only if

$$\max_{\Delta : |\Delta| \leq \rho} 2 [[P^T Z^T R \xi]^T \Delta [Q \xi]] \leq \xi^T Y \xi \quad \forall \xi \in \mathbf{R}^n. \quad (\text{Property 2})$$

¹¹⁾ This indeed is a “shrinkage”: $|\Delta| \leq \|\Delta\|_2$ for every matrix Δ (prove it!)

The maximum over Δ , $|\Delta| \leq \rho$, of the quantity $\eta^T \Delta \zeta$, clearly is equal to ρ times the product of the Euclidean norms of the vectors η and ζ (why?). Thus, Property 2 is equivalent to

$$\xi^T Y \xi - 2\rho \|Q\xi\|_2 \|P^T Z^T R \xi\|_2 \geq 0 \quad \forall \xi \in \mathbf{R}^n. \quad (\text{Property 3})$$

Now is the trick: Property 3 is clearly equivalent to the following

Property 4: Every pair $\zeta = (\xi, \eta) \in \mathbf{R}^n \times \mathbf{R}^k$ which satisfies the quadratic inequality

$$\xi^T Q^T Q \xi - \eta^T \eta \geq 0 \quad (\text{I})$$

satisfies also the quadratic inequality

$$\xi^T Y \xi - 2\rho \eta^T P^T Z^T R \xi \geq 0. \quad (\text{II})$$

Indeed, for a fixed ξ the minimum over η satisfying (I) of the left hand side in (II) is nothing but the left hand side in (Property 3).

It remains to use the \mathcal{S} -Lemma:

\mathcal{S} -Lemma. Let A, B be symmetric $n \times n$ matrices, and assume that the quadratic inequality

$$x^T A x \geq 0 \quad (\text{A})$$

is strictly feasible: there exists \bar{x} such that $\bar{x}^T A \bar{x} > 0$. Then the quadratic inequality

$$x^T B x \geq 0 \quad (\text{B})$$

is a consequence of (A) if and only if it is a linear consequence of (A), i.e., if and only if there exists a nonnegative λ such that

$$B \succeq \lambda A.$$

(for a proof, see Section 3.5). Property 4 says that the quadratic inequality (II) with variables ξ, η is a consequence of (I); by the \mathcal{S} -Lemma (recall that $Q \neq 0$, so that (I) is strictly feasible!) this is equivalent to the existence of a nonnegative λ such that

$$\begin{pmatrix} Y & -\rho R^T Z P \\ -\rho P^T Z^T R & \end{pmatrix} - \lambda \begin{pmatrix} Q^T Q & \\ & -I_k \end{pmatrix} \succeq 0,$$

which is exactly the statement of Lemma 3.3.2 for the case of $|\Delta| \leq \rho$. The case of perturbations with $\|\Delta\|_2 \leq \rho$ is completely similar, since the equivalence between Properties 2 and 3 is valid independently of which norm of $\Delta - |\cdot|$ or $\|\cdot\|_2$ is used.

3.3.3.3 Lyapunov Stability Synthesis

We have seen that under reasonable assumptions on the underlying uncertainty set the question of whether a given uncertain linear system (ULS) admits a quadratic Lyapunov function can be reduced to a semidefinite program. Now let us switch from the *analysis* question: “whether a stability of an uncertain linear system may be certified by a quadratic Lyapunov function” to the *synthesis* question which is as follows. Assume that we are given an *uncertain open loop controlled system*

$$\begin{aligned} \frac{d}{dt} x(t) &= A(t)x(t) + B(t)u(t) \\ y(t) &= C(t)x(t); \end{aligned} \quad (\text{UOS})$$

all we know about the collection $(A(t), B(t), C(t))$ of time-varying $n \times n$ matrix $A(t)$, $n \times k$ matrix $B(t)$ and $l \times n$ matrix $C(t)$ is that this collection, at every time t , belongs to a given

uncertainty set \mathcal{U} . The question is whether we can equip our uncertain “open loop” system (UOS) with a linear feedback

$$u(t) = Ky(t)$$

in such a way that the resulting uncertain closed loop system

$$\frac{d}{dt}x(t) = [A(t) + B(t)KC(t)]x(t) \quad (\text{UCS})$$

will be stable and, moreover, such that its stability can be certified by a quadratic Lyapunov function. In other words, now we are simultaneously looking for a “stabilizing controller” and a quadratic Lyapunov certificate of its stabilizing ability.

With the “global linearization” trick we may use the results on uncertain controlled linear systems to build stabilizing linear controllers for *nonlinear* controlled systems

$$\begin{aligned} \frac{d}{dt}x(t) &= f(t, x(t), u(t)) \\ y(t) &= g(t, x(t)) \end{aligned}$$

Assuming $f(t, 0, 0) = 0$, $g(t, 0) = 0$ and denoting by \mathcal{U} the closed convex hull of the set

$$\left\{ \left(\frac{\partial}{\partial x}f(t, x, u), \frac{\partial}{\partial u}f(t, x, u), \frac{\partial}{\partial x}g(t, x) \right) \mid t \geq 0, x \in \mathbf{R}^n, u \in \mathbf{R}^k \right\},$$

we see that every trajectory of the original nonlinear system is a trajectory of the uncertain linear system (UOS) associated with the set \mathcal{U} . Consequently, if we are able to find a stabilizing controller for (UOS) and certify its stabilizing property by a quadratic Lyapunov function, then the resulting controller/Lyapunov function will stabilize the nonlinear system and will certify the stability of the closed loop system, respectively.

Exactly the same reasoning as in the previous Section leads us to the following

Proposition 3.3.3 *Let \mathcal{U} be the uncertainty set associated with an uncertain open loop controlled system (UOS). The system admits a stabilizing controller along with a quadratic Lyapunov stability certificate for the resulting closed loop system if and only if the optimal value in the optimization problem*

$$\begin{aligned} \text{minimize} \quad & s \\ \text{s.t.} \quad & [A + BKC]^T X + X[A + BKC] \preceq sI_n \quad \forall(A, B, C) \in \mathcal{U} \\ & X \succeq I_n, \end{aligned} \quad (\text{LyS})$$

in design variables s, X, K , is negative. Moreover, every feasible solution to the problem with negative value of the objective provides stabilizing controller along with quadratic Lyapunov stability certificate for the resulting closed loop system.

A bad news about (LyS) is that it is much more difficult to rewrite this problem as a semidefinite program than in the analysis case (i.e., the case of $K = 0$), since (LyS) is a semi-infinite system of *nonlinear* matrix inequalities. There is, however, an important particular case where this difficulty can be eliminated. This is the case of a feedback via the *full* state vector – the case when $y(t) = x(t)$ (i.e., $C(t)$ is the unit matrix). In this case, all we need in order to get a

stabilizing controller along with a quadratic Lyapunov certificate of its stabilizing ability, is to solve a system of *strict* matrix inequalities

$$\begin{array}{ccc} [A + BK]^T X + X[A + BK] & \preceq & Z \prec 0 \\ X & \succ & 0 \end{array} \quad \forall(A, B) \in \mathcal{U}. \quad (*)$$

Indeed, given a solution (X, K, Z) to this system, we always can convert it by normalization of X to a solution of (LyS). Now let us make the change of variables

$$Y = X^{-1}, L = KX^{-1}, W = X^{-1}ZX^{-1} \quad [\Leftrightarrow X = Y^{-1}, K = LY^{-1}, Z = Y^{-1}WY^{-1}].$$

With respect to the new variables Y, L, K system (*) becomes

$$\begin{cases} [A + BLY^{-1}]^T Y^{-1} + Y^{-1}[A + BLY^{-1}] & \preceq Y^{-1}WY^{-1} \prec 0 \\ Y & \succ 0 \\ \Updownarrow \\ \begin{cases} L^T B^T + YA^T + BL + AY & \preceq W \prec 0, \quad \forall(A, B) \in \mathcal{U} \\ Y & \succ 0 \end{cases} \end{cases}$$

(we have multiplied all original matrix inequalities from the left and from the right by Y). What we end up with, is a system of strict *linear* matrix inequalities with respect to our new design variables L, Y, W ; the question of whether this system is solvable can be converted to the question of whether the optimal value in a problem of the type (LyS) is negative, and we come to the following

Proposition 3.3.4 *Consider an uncertain controlled linear system with a full observer:*

$$\begin{array}{lcl} \frac{d}{dt}x(t) & = & A(t)x(t) + B(t)u(t) \\ y(t) & = & x(t) \end{array}$$

and let \mathcal{U} be the corresponding uncertainty set (which now is comprised of pairs (A, B) of possible values of $(A(t), B(t))$, since $C(t) \equiv I_n$ is certain).

The system can be stabilized by a linear controller

$$u(t) = Ky(t) \quad [\equiv Kx(t)]$$

in such a way that the resulting uncertain closed loop system

$$\frac{d}{dt}x(t) = [A(t) + B(t)K]x(t)$$

admits a quadratic Lyapunov stability certificate if and only if the optimal value in the optimization problem

$$\begin{array}{ll} \text{minimize} & s \\ \text{s.t.} & \\ & BL + AY + L^T B^T + YA^T \preceq sI_n \quad \forall(A, B) \in \mathcal{U} \\ & Y \succeq I \end{array} \quad (\text{Ly}^*)$$

in the design variables $s \in \mathbf{R}$, $Y \in \mathbf{S}^n$, $L \in \mathbf{M}^{k,n}$, is negative. Moreover, every feasible solution to (Ly^{*}) with negative value of the objective provides a stabilizing linear controller along with related quadratic Lyapunov stability certificate.

In particular, in the polytopic case:

$$\mathcal{U} = \text{Conv}\{(A_1, B_1), \dots, (A_N, B_N)\}$$

the Quadratic Lyapunov Stability Synthesis reduces to solving the semidefinite program

$$\min_{s, Y, L} \left\{ s : B_i L + A_i Y + Y A_i^T + L^T B_i^T \preceq s I_n, i = 1, \dots, N; Y \succeq I_n \right\}.$$

3.4 Semidefinite relaxations of intractable problems

One of the most challenging and promising applications of Semidefinite Programming is in building tractable approximations of “computationally intractable” optimization problems. Let us look at several applications of this type.

3.4.1 Semidefinite relaxations of combinatorial problems

3.4.1.1 Combinatorial problems and their relaxations

Numerous problems of planning, scheduling, routing, etc., can be posed as *combinatorial optimization problems*, i.e., optimization programs with discrete design variables (integer or zero-one). There are several “universal forms” of combinatorial problems, among them Linear Programming with integer variables and Linear Programming with 0-1 variables; a problem given in one of these forms can always be converted to any other universal form, so that in principle it does not matter which form to use. Now, the majority of combinatorial problems are difficult – we do not know theoretically efficient (in certain precise meaning of the notion) algorithms for solving these problems. What we do know is that nearly all these difficult problems are, in a sense, equivalent to each other and are *NP-complete*. The exact meaning of the latter notion will be explained in Lecture 4; for the time being it suffices to say that NP-completeness of a problem P means that the problem is “as difficult as a combinatorial problem can be” – if we knew an efficient algorithm for P , we would be able to convert it to an efficient algorithm for any other combinatorial problem. NP-complete problems may look extremely “simple”, as it is demonstrated by the following example:

(Stones) Given n stones of positive integer weights (i.e., given n positive integers a_1, \dots, a_n), check whether you can partition these stones into two groups of equal weight, i.e., check whether a linear equation

$$\sum_{i=1}^n a_i x_i = 0$$

has a solution with $x_i = \pm 1$.

Theoretically difficult combinatorial problems happen to be difficult to solve in practice as well. An important ingredient in basically all algorithms for combinatorial optimization is a technique for building bounds for the unknown optimal value of a given (sub)problem. A typical way to estimate the optimal value of an optimization program

$$f^* = \min_x \{f(x) : x \in X\}$$

from above is to present a feasible solution \bar{x} ; then clearly $f^* \leq f(\bar{x})$. And a typical way to bound the optimal value from below is to pass from the problem to its *relaxation*

$$f_* = \min_x \{f(x) : x \in X'\}$$

increasing the feasible set: $X \subset X'$. Clearly, $f_* \leq f^*$, so, whenever the relaxation is efficiently solvable (to ensure this, we should take care of how we choose X'), it provides us with a “computable” lower bound on the actual optimal value.

When building a relaxation, one should take care of two issues: on one hand, we want the relaxation to be “efficiently solvable”. On the other hand, we want the relaxation to be “tight”, otherwise the lower bound we get may be by far “too optimistic” and therefore not useful. For a long time, the only practical relaxations were the LP ones, since these were the only problems one could solve efficiently. With recent progress in optimization techniques, nonlinear relaxations become more and more “practical”; as a result, we are witnessing a growing theoretical and computational activity in the area of nonlinear relaxations of combinatorial problems. These developments mostly deal with *semidefinite relaxations*. Let us look how they emerge.

3.4.1.2 Shor's Semidefinite Relaxation scheme

As it was already mentioned, there are numerous “universal forms” of combinatorial problems. E.g., a combinatorial problem can be posed as minimizing a quadratic objective under quadratic inequality constraints:

$$\begin{aligned} & \text{minimize in } x \in \mathbf{R}^n \quad f_0(x) = x^T A_0 x + 2b_0^T x + c_0 \\ & \text{s.t.} \\ & \quad f_i(x) = x^T A_i x + 2b_i^T x + c_i \leq 0, \quad i = 1, \dots, m. \end{aligned} \tag{3.4.1}$$

To see that this form is “universal”, note that it covers the classical universal combinatorial problem – a generic LP program with Boolean (0-1) variables:

$$\min_x \left\{ c^T x : a_i^T x - b_i \leq 0, \quad i = 1, \dots, m; x_j \in \{0, 1\}, \quad j = 1, \dots, n \right\} \tag{B}$$

Indeed, the fact that a variable x_j must be Boolean can be expressed by the quadratic equality

$$x_j^2 - x_j = 0,$$

which can be represented by a pair of opposite quadratic inequalities and a linear inequality $a_i^T x - b_i \leq 0$ is a particular case of quadratic inequality. Thus, (B) is equivalent to the problem

$$\min_{x,s} \left\{ c^T x : a_i^T x - b_i \leq 0, \quad i = 1, \dots, m; x_j^2 - x_j \leq 0, \quad -x_j^2 + x_j \leq 0, \quad j = 1, \dots, n \right\},$$

and this problem is of the form (3.4.1).

To bound from below the optimal value in (3.4.1), we may use the same technique we used for building the dual problem (it is called the *Lagrange relaxation*). We choose somehow “weights” $\lambda_i \geq 0$, $i = 1, \dots, m$, and add the constraints of (3.4.1) with these weights to the objective, thus coming to the function

$$\begin{aligned} f_\lambda(x) &= f_0(x) + \sum_{i=1}^m \lambda_i f_i(x) \\ &= x^T A(\lambda) x + 2b^T(\lambda) x + c(\lambda), \end{aligned} \tag{3.4.2}$$

where

$$\begin{aligned} A(\lambda) &= A_0 + \sum_{i=1}^m \lambda_i A_i \\ b(\lambda) &= b_0 + \sum_{i=1}^m \lambda_i b_i \\ c(\lambda) &= c_0 + \sum_{i=1}^m \lambda_i c_i \end{aligned}$$

By construction, the function $f_\lambda(x)$ is \leq the actual objective $f_0(x)$ on the feasible set of the problem (3.4.1). Consequently, the unconstrained infimum of this function

$$a(\lambda) = \inf_{x \in \mathbf{R}^n} f_\lambda(x)$$

is a lower bound for the optimal value in (3.4.1). We come to the following simple result (cf. the Weak Duality Theorem):

(*) Assume that $\lambda \in \mathbf{R}_+^m$ and $\zeta \in \mathbf{R}$ are such that

$$f_\lambda(x) - \zeta \geq 0 \quad \forall x \in \mathbf{R}^n \quad (3.4.3)$$

(i.e., that $\zeta \leq a(\lambda)$). Then ζ is a lower bound for the optimal value in (3.4.1).

It remains to clarify what does it mean that (3.4.3) holds. Recalling the structure of f_λ , we see that it means that the inhomogeneous quadratic form

$$g_\lambda(x) = x^T A(\lambda)x + 2b^T(\lambda)x + c(\lambda) - \zeta$$

is nonnegative on the entire space. Now, an inhomogeneous quadratic form

$$g(x) = x^T Ax + 2b^T x + c$$

is nonnegative everywhere if and only if certain associated *homogeneous* quadratic form is nonnegative everywhere. Indeed, given $t \neq 0$ and $x \in \mathbf{R}^n$, the fact that $g(t^{-1}x) \geq 0$ means exactly the nonnegativity of the *homogeneous* quadratic form $G(x, t)$

$$G(x, t) = x^T Ax + 2tb^T x + ct^2$$

with $(n + 1)$ variables x, t . We see that if g is nonnegative, then G is nonnegative whenever $t \neq 0$; by continuity, G then is nonnegative everywhere. Thus, if g is nonnegative, then G is, and of course vice versa (since $g(x) = G(x, 1)$). Now, to say that G is nonnegative everywhere is literally the same as to say that the matrix

$$\begin{pmatrix} c & b^T \\ b & A \end{pmatrix} \quad (3.4.4)$$

is positive semidefinite.

It is worthy to catalogue our simple observation:

Simple Lemma. A quadratic inequality with a (symmetric) $n \times n$ matrix A

$$x^T Ax + 2b^T x + c \geq 0$$

is identically true – is valid for all $x \in \mathbf{R}^n$ – if and only if the matrix (3.4.4) is positive semidefinite.

Applying this observation to $g_\lambda(x)$, we get the following equivalent reformulation of (*):

If $(\lambda, \zeta) \in \mathbf{R}_+^m \times \mathbf{R}$ satisfy the LMI

$$\begin{pmatrix} \sum_{i=1}^m \lambda_i c_i - \zeta & b_0^T + \sum_{i=1}^m \lambda_i b_i^T \\ b_0 + \sum_{i=1}^m \lambda_i b_i & A_0 + \sum_{i=1}^m \lambda_i A_i \end{pmatrix} \succeq 0,$$

then ζ is a lower bound for the optimal value in (3.4.1).

Now, what is the best lower bound we can get with this scheme? Of course, it is the optimal value of the semidefinite program

$$\max_{\zeta, \lambda} \left\{ \zeta : \begin{pmatrix} c_0 + \sum_{i=1}^m \lambda_i c_i - \zeta & b_0^T + \sum_{i=1}^m \lambda_i b_i^T \\ b_0 + \sum_{i=1}^m \lambda_i b_i & A_0 + \sum_{i=1}^m \lambda_i A_i \end{pmatrix} \succeq 0, \lambda \geq 0 \right\}. \quad (3.4.5)$$

We have proved the following simple

Proposition 3.4.1 *The optimal value in (3.4.5) is a lower bound for the optimal value in (3.4.1).*

The outlined scheme is extremely transparent, but it looks different from a relaxation scheme as explained above – where is the extension of the feasible set of the original problem? In fact the scheme is of this type. To see it, note that the value of a quadratic form at a point $x \in \mathbf{R}^n$ can be written as the Frobenius inner product of a matrix defined by the problem data and the dyadic matrix $X(x) = \begin{pmatrix} 1 \\ x \end{pmatrix} \begin{pmatrix} 1 \\ x \end{pmatrix}^T$:

$$x^T Ax + 2b^T x + c = \begin{pmatrix} 1 \\ x \end{pmatrix}^T \begin{pmatrix} c & b^T \\ b & A \end{pmatrix} \begin{pmatrix} 1 \\ x \end{pmatrix} = \text{Tr} \left(\begin{pmatrix} c & b^T \\ b & A \end{pmatrix} X(x) \right).$$

Consequently, (3.4.1) can be written as

$$\min_x \left\{ \text{Tr} \left(\begin{pmatrix} c_0 & b_0^T \\ b_0 & A_0 \end{pmatrix} X(x) \right) : \text{Tr} \left(\begin{pmatrix} c_i & b_i^T \\ b_i & A_i \end{pmatrix} X(x) \right) \leq 0, i = 1, \dots, m \right\}. \quad (3.4.6)$$

Thus, we may think of (3.4.2) as a problem with *linear* objective and *linear* equality constraints and with the design vector X which is a symmetric $(n+1) \times (n+1)$ matrix running through the nonlinear manifold \mathcal{X} of dyadic matrices $X(x)$, $x \in \mathbf{R}^n$. Clearly, all points of \mathcal{X} are positive semidefinite matrices with North-Western entry 1. Now let $\bar{\mathcal{X}}$ be the set of all such matrices. Replacing \mathcal{X} by $\bar{\mathcal{X}}$, we get a relaxation of (3.4.6) (the latter problem is, essentially, our original problem (3.4.1)). This relaxation is the semidefinite program

$$\begin{aligned} \min_X \{ \text{Tr}(\bar{A}_0 X) : \text{Tr}(\bar{A}_i X) \leq 0, i = 1, \dots, m; X \succeq 0; X_{11} = 1 \} \\ \left[A_i = \begin{pmatrix} c_i & b_i^T \\ b_i & A_i \end{pmatrix}, i = 1, \dots, m \right]. \end{aligned} \quad (3.4.7)$$

We get the following

Proposition 3.4.2 *The optimal value of the semidefinite program (3.4.7) is a lower bound for the optimal value in (3.4.1).*

One can easily verify that problem (3.4.5) is just the semidefinite dual of (3.4.7); thus, when deriving (3.4.5), we were in fact implementing the idea of relaxation. This is why in the sequel we call both (3.4.7) and (3.4.5) *semidefinite relaxations* of (3.4.1).

3.4.1.3 When the semidefinite relaxation is exact?

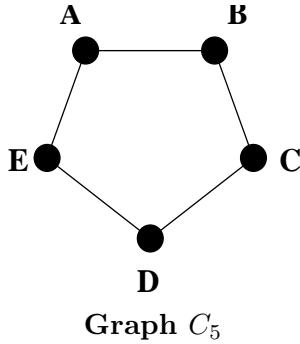
In general, the optimal value in (3.4.7) is just a lower bound on the optimal value of (3.4.1). There are, however, two cases when this bound is exact. These are

- Convex case, where all quadratic forms in (3.4.1) are convex (i.e., $Q_i \succeq 0$, $i = 0, 1, \dots, m$). Here, strict feasibility of the problem (i.e., the existence of a feasible solution \bar{x} with $f_i(\bar{x}) < 0$, $i = 1, \dots, m$) plus below boundedness of it imply that (3.4.7) is solvable with the optimal value equal to the one of (3.4.1). This statement is a particular case of the well-known *Lagrange Duality Theorem* in Convex Programming.
- The case of $m = 1$. Here the optimal value in (3.4.1) is equal to the one in (3.4.7), provided that (3.4.1) is strictly feasible. This highly surprising fact (no convexity is assumed!) is called *Inhomogeneous S-Lemma*; we shall prove it in Section .

Let us look at several interesting examples of Semidefinite relaxations.

3.4.1.4 Stability number, Shannon capacity and Lovasz capacity of a graph

Stability number of a graph. Consider a (non-oriented) graph – a finite set of nodes linked by arcs¹²⁾, like the simple 5-node graph C_5 :



One of the fundamental characteristics of a graph Γ is its *stability number* $\alpha(\Gamma)$ defined as the maximum cardinality of an *independent* subset of nodes – a subset such that no two nodes from it are linked by an arc. E.g., the stability number for the graph C_5 is 2, and a maximal independent set is, e.g., $\{A; C\}$.

The problem of computing the stability number of a given graph is NP-complete, this is why it is important to know how to bound this number.

Shannon capacity of a graph. An upper bound on the stability number of a graph which interesting by its own right is the *Shannon capacity* $\Theta(\Gamma)$ defined as follows.

Let us treat the nodes of Γ as letters of certain alphabet, and the arcs as possible errors in certain communication channel: you can send through the channel one letter per unit time, and what arrives on the other end of the channel can be either the letter you have sent, or any letter adjacent to it. Now assume that you are planning to communicate with an addressee through the

¹²⁾One of the formal definitions of a (non-oriented) graph is as follows: a n -node graph is just a $n \times n$ symmetric matrix A with entries 0, 1 and zero diagonal. The rows (and the columns) of the matrix are identified with the nodes $1, 2, \dots, n$ of the graph, and the nodes i, j are adjacent (i.e., linked by an arc) exactly for those i, j with $A_{ij} = 1$.

channel by sending n -letter words (n is fixed). You fix in advance a dictionary D_n of words to be used and make this dictionary known to the addressee. What you are interested in when building the dictionary is to get a good one, meaning that no word from it could be transformed by the channel into another word from the dictionary. If your dictionary satisfies this requirement, you may be sure that the addressee will never misunderstand you: whatever word from the dictionary you send and whatever possible transmission errors occur, the addressee is able either to get the correct message, or to realize that the message was corrupted during transmission, but there is no risk that your “yes” will be read as “no!”. Now, in order to utilize the channel “at full capacity”, you are interested to get as large dictionary as possible. How many words it can include? The answer is clear: this is precisely the stability number of the graph Γ^n as follows: the nodes of Γ^n are ordered n -element collections of the nodes of Γ – all possible n -letter words in your alphabet; two distinct nodes (i_1, \dots, i_n) (j_1, \dots, j_n) are adjacent in Γ^n if and only if for every l the l -th letters i_l and j_l in the two words either coincide, or are adjacent in Γ (i.e., two distinct n -letter words are adjacent, if the transmission can convert one of them into the other one). Let us denote the maximum number of words in a “good” dictionary D_n (i.e., the stability number of Γ^n) by $f(n)$, The function $f(n)$ possesses the following nice property:

$$f(k)f(l) \leq f(k+l), \quad k, l = 1, 2, \dots \quad (*)$$

Indeed, given the best (of the cardinality $f(k)$) good dictionary D_k and the best good dictionary D_l , let us build a dictionary comprised of all $(k+l)$ -letter words as follows: the initial k -letter fragment of a word belongs to D_k , and the remaining l -letter fragment belongs to D_l . The resulting dictionary is clearly good and contains $f(k)f(l)$ words, and $(*)$ follows.

Now, this is a simple exercise in analysis to see that for a nonnegative function f with property $(*)$ one has

$$\lim_{k \rightarrow \infty} (f(k))^{1/k} = \sup_{k \geq 1} (f(k))^{1/k} \in [0, +\infty].$$

In our situation $\sup_{k \geq 1} (f(k))^{1/k} < \infty$, since clearly $f(k) \leq n^k$, n being the number of letters (the number of nodes in Γ). Consequently, the quantity

$$\Theta(\Gamma) = \lim_{k \rightarrow \infty} (f(k))^{1/k}$$

is well-defined; moreover, for every k the quantity $(f(k))^{1/k}$ is a lower bound for $\Theta(\Gamma)$. The number $\Theta(\Gamma)$ is called the *Shannon capacity* of Γ . Our immediate observation is that

(!) *The Shannon capacity $\Theta(\Gamma)$ majorates the stability number of Γ :*

$$\alpha(\Gamma) \leq \Theta(\Gamma).$$

Indeed, as we remember, $(f(k))^{1/k}$ is a lower bound for $\Theta(\Gamma)$ for every $k = 1, 2, \dots$; setting $k = 1$ and taking into account that $f(1) = \alpha(\Gamma)$, we get the desired result.

We see that the Shannon capacity number is an upper bound on the stability number; and this bound has a nice interpretation in terms of the Information Theory. The bad news is that we do not know how to compute the Shannon capacity. E.g., what is it for the toy graph C_5 ?

The stability number of C_5 clearly is 2, so that our first observation is that

$$\Theta(C_5) \geq \alpha(C_5) = 2.$$

To get a better estimate, let us look the graph $(C_5)^2$ (as we remember, $\Theta(\Gamma) \geq (f(k))^{1/k} = (\alpha(\Gamma^k))^{1/k}$ for every k). The graph $(C_5)^2$ has 25 nodes, so that we do not draw it; it, however, is not that difficult to find its stability number, which turns out to be 5. A good 5-element dictionary (\equiv a 5-node independent set in $(C_5)^2$) is, e.g.,

$$AA, BC, CE, DB, ED.$$

Thus, we get

$$\Theta(C_5) \geq \sqrt{\alpha((C_5)^2)} = \sqrt{5}.$$

Attempts to compute the subsequent lower bounds $(f(k))^{1/k}$, as long as they are implementable (think how many vertices there are in $(C_5)^4!$), do not yield any improvements, and for more than 20 years it remained unknown whether $\Theta(C_5) = \sqrt{5}$ or is $> \sqrt{5}$. And this is for a toy graph! The breakthrough in the area of upper bounds for the stability number is due to L. Lovasz who in early 70's found a new – computable! – bound of this type.

Lovasz capacity number. Given a n -node graph Γ , let us associate with it an affine matrix-valued function $\mathcal{L}(x)$ taking values in the space of $n \times n$ symmetric matrices, namely, as follows:

- For every pair i, j of indices ($1 \leq i, j \leq n$) such that the nodes i and j are not linked by an arc, the ij -th entry of \mathcal{L} is equal to 1;
- For a pair $i < j$ of indices such that the nodes i, j are linked by an arc, the ij -th and the ji -th entries in \mathcal{L} are equal to x_{ij} – to the variable associated with the arc (i, j) .

Thus, $\mathcal{L}(x)$ is indeed an affine function of N design variables x_{ij} , where N is the number of arcs in the graph. E.g., for graph C_5 the function \mathcal{L} is as follows:

$$\mathcal{L} = \begin{pmatrix} 1 & x_{AB} & 1 & 1 & x_{EA} \\ x_{AB} & 1 & x_{BC} & 1 & 1 \\ 1 & x_{BC} & 1 & x_{CD} & 1 \\ 1 & 1 & x_{CD} & 1 & x_{DE} \\ x_{EA} & 1 & 1 & x_{DE} & 1 \end{pmatrix}.$$

Now, the Lovasz capacity number $\vartheta(\Gamma)$ is defined as the optimal value of the optimization program

$$\min_x \{\lambda_{\max}(\mathcal{L}(x))\},$$

i.e., as the optimal value in the semidefinite program

$$\min_{\lambda, x} \{\lambda : \lambda I_n - \mathcal{L}(x) \succeq 0\}. \quad (\text{L})$$

Proposition 3.4.3 [Lovasz] *The Lovasz capacity number is an upper bound for the Shannon capacity:*

$$\vartheta(\Gamma) \geq \Theta(\Gamma)$$

and, consequently, for the stability number:

$$\vartheta(\Gamma) \geq \Theta(\Gamma) \geq \alpha(\Gamma).$$

For the graph C_5 , the Lovasz capacity can be easily computed analytically and turns out to be exactly $\sqrt{5}$. Thus, a small byproduct of Lovasz's result is a solution to the problem which remained open for two decades.

Let us look how the Lovasz bound on the stability number can be obtained from the general relaxation scheme. To this end note that the stability number of an n -node graph Γ is the optimal value of the following optimization problem with 0-1 variables:

$$\begin{aligned} \max_x & \left\{ e^T x : x_i x_j = 0 \text{ whenever } i, j \text{ are adjacent nodes, } x_i \in \{0, 1\}, i = 1, \dots, n \right\}, \\ & e = (1, \dots, 1)^T \in \mathbf{R}^n. \end{aligned}$$

Indeed, 0-1 n -dimensional vectors can be identified with sets of nodes of Γ : the coordinates x_i of the vector x representing a set A of nodes are ones for $i \in A$ and zeros otherwise. The quadratic equality constraints $x_i x_j = 0$ for such a vector express equivalently the fact that the corresponding set of nodes is independent, and the objective $e^T x$ counts the cardinality of this set.

As we remember, the 0-1 restrictions on the variables can be represented equivalently by quadratic equality constraints, so that the stability number of Γ is the optimal value of the following problem with quadratic (in fact linear) objective and quadratic equality constraints:

$$\begin{aligned} \text{maximize} \quad & e^T x \\ \text{s.t.} \quad & x_i x_j = 0, (i, j) \text{ is an arc} \\ & x_i^2 - x_i = 0, i = 1, \dots, n. \end{aligned} \tag{3.4.8}$$

The latter problem is in the form of (3.4.1), with the only difference that the objective should be maximized rather than minimized. Switching from maximization of $e^T x$ to minimization of $(-e)^T x$ and passing to (3.4.5), we get the problem

$$\max_{\zeta, \mu} \left\{ \zeta : \begin{pmatrix} -\zeta & -\frac{1}{2}(e + \mu)^T \\ -\frac{1}{2}(e + \mu) & A(\mu, \lambda) \end{pmatrix} \succeq 0 \right\},$$

where μ is n -dimensional and $A(\mu, \lambda)$ is as follows:

- The diagonal entries of $A(\mu, \lambda)$ are μ_1, \dots, μ_n ;
- The off-diagonal cells ij corresponding to non-adjacent nodes i, j ("empty cells") are zeros;
- The off-diagonal cells ij , $i < j$, and the symmetric cells ji corresponding to adjacent nodes i, j ("arc cells") are filled with free variables λ_{ij} .

Note that the optimal value in the resulting problem is a *lower bound* for *minus* the optimal value of (3.4.8), i.e., for minus the stability number of Γ .

Passing in the resulting problem from the variable ζ to a new variable $\xi = -\zeta$ and again switching from maximization of $\zeta = -\xi$ to minimization of ξ , we end up with the semidefinite program

$$\min_{\xi, \lambda, \mu} \left\{ \xi : \begin{pmatrix} \xi & -\frac{1}{2}(e + \mu)^T \\ -\frac{1}{2}(e + \mu) & A(\mu, \lambda) \end{pmatrix} \succeq 0 \right\}. \tag{3.4.9}$$

The optimal value in this problem is the minus optimal value in the previous one, which, in turn, is a lower bound on the minus stability number of Γ ; consequently, the optimal value in (3.4.9) is an *upper bound* on the stability number of Γ .

We have built a semidefinite relaxation (3.4.9) of the problem of computing the stability number of Γ ; the optimal value in the relaxation is an upper bound on the stability number. To get the Lovasz relaxation, let us further fix the μ -variables at the level 1 (this may only increase the optimal value in the problem, so that it still will be an upper bound for the stability number)¹³⁾. With this modification, we come to the problem

$$\min_{\xi, \lambda} \left\{ \xi : \begin{pmatrix} \xi & -e^T \\ -e & A(e, \lambda) \end{pmatrix} \succeq 0 \right\}.$$

In every feasible solution to the problem, ξ should be ≥ 1 (it is an upper bound for $\alpha(\Gamma) \geq 1$). When $\xi \geq 1$, the LMI

$$\begin{pmatrix} \xi & -e^T \\ e & A(e, \lambda) \end{pmatrix} \succeq 0$$

by the Schur Complement Lemma is equivalent to the LMI

$$A(e, \lambda) \succeq (-e)\xi^{-1}(-e)^T,$$

or, which is the same, to the LMI

$$\xi A(e, \lambda) - ee^T \succeq 0.$$

The left hand side matrix in the latter LMI is equal to $\xi I_n - B(\xi, \lambda)$, where the matrix $B(\xi, \lambda)$ is as follows:

- The diagonal entries of $B(\xi, \lambda)$ are equal to 1;
- The off-diagonal “empty cells” are filled with ones;
- The “arc cells” from a symmetric pair off-diagonal pair ij and ji ($i < j$) are filled with $\xi \lambda_{ij}$.

Passing from the design variables λ to the new ones $x_{ij} = \xi \lambda_{ij}$, we conclude that problem (3.4.9) with μ 's set to ones is equivalent to the problem

$$\min_{\xi, x} \{ \xi \rightarrow \min \mid \xi I_n - \mathcal{L}(x) \succeq 0 \},$$

whose optimal value is exactly the Lovasz capacity number of Γ .

As a byproduct of our derivation, we get the easy part of the Lovasz Theorem – the inequality $\vartheta(\Gamma) \geq \alpha(\Gamma)$; this inequality, however, could be easily obtained directly from the definition of $\vartheta(\Gamma)$. The advantage of our derivation is that it demonstrates what is the origin of $\vartheta(\Gamma)$.

How good is the Lovasz capacity number? The Lovasz capacity number plays a crucial role in numerous graph-related problems; there is an important sub-family of graphs – *perfect graphs* – for which this number coincides with the stability number. However, for a general-type graph Γ , $\vartheta(\Gamma)$ may be a fairly poor bound for $\alpha(\Gamma)$. Lovasz has proved that for any graph Γ with n nodes, $\vartheta(\Gamma)\vartheta(\hat{\Gamma}) \geq n$, where $\hat{\Gamma}$ is the *complement* to Γ (i.e., two distinct nodes are adjacent in $\hat{\Gamma}$ if and only if they are *not* adjacent in Γ). It follows that for n -node graph Γ one always has $\max[\vartheta(\Gamma), \vartheta(\hat{\Gamma})] \geq \sqrt{n}$. On the other hand, it turns out that for a random n -node graph Γ (the arcs are drawn at random and independently of each other, with probability 0.5 to draw an arc

¹³⁾In fact setting $\mu_i = 1$, we do not vary the optimal value at all.

linking two given distinct nodes) $\max[\alpha(\Gamma), \alpha(\hat{\Gamma})]$ is “typically” (with probability approaching 1 as n grows) of order of $\ln n$. It follows that for random n -node graphs a typical value of the ratio $\vartheta(\Gamma)/\alpha(\Gamma)$ is at least of order of $n^{1/2}/\ln n$; as n grows, this ratio blows up to ∞ .

A natural question arises: are there “difficult” (NP-complete) combinatorial problems admitting “good” semidefinite relaxations – those with the quality of approximation not deteriorating as the sizes of instances grow? Let us look at two breakthrough results in this direction.

3.4.1.5 The MAXCUT problem and maximizing quadratic form over a box

The MAXCUT problem. The *maximum cut* problem is as follows:

Problem 3.4.1 [MAXCUT] *Let Γ be an n -node graph, and let the arcs (i, j) of the graph be associated with nonnegative “weights” a_{ij} . The problem is to find a cut of the largest possible weight, i.e., to partition the set of nodes in two parts S, S' in such a way that the total weight of all arcs “linking S and S' ” (i.e., with one incident node in S and the other one in S') is as large as possible.*

In the MAXCUT problem, we may assume that the weights $a_{ij} = a_{ji} \geq 0$ are defined for every pair i, j of indices; it suffices to set $a_{ij} = 0$ for pairs i, j of non-adjacent nodes.

In contrast to the *minimum cut* problem (where we should minimize the weight of a cut instead of maximizing it), which is, basically, a nice LP program of finding the maximum flow in a network and is therefore efficiently solvable, the MAXCUT problem is as difficult as a combinatorial problem can be – it is NP-complete.

Theorem of Goemans and Williamson [12]. It is easy to build a semidefinite relaxation of MAXCUT. To this end let us pose MAXCUT as a quadratic problem with quadratic equality constraints. Let Γ be a n -node graph. A cut (S, S') – a partitioning of the set of nodes in two disjoint parts S, S' – can be identified with a n -dimensional vector x with coordinates ± 1 – $x_i = 1$ for $i \in S$, $x_i = -1$ for $i \in S'$. The quantity $\frac{1}{2} \sum_{i,j=1}^n a_{ij}x_i x_j$ is the total weight of arcs with both ends either in S or in S' minus the weight of the cut (S, S') ; consequently, the quantity

$$\frac{1}{2} \left[\frac{1}{2} \sum_{i,j=1}^n a_{ij} - \frac{1}{2} \sum_{i,j=1}^n a_{ij}x_i x_j \right] = \frac{1}{4} \sum_{i,j=1}^n a_{ij}(1 - x_i x_j)$$

is exactly the weight of the cut (S, S') .

We conclude that the MAXCUT problem can be posed as the following quadratic problem with quadratic equality constraints:

$$\max_x \left\{ \frac{1}{4} \sum_{i,j=1}^n a_{ij}(1 - x_i x_j) : x_i^2 = 1, i = 1, \dots, n \right\}. \quad (3.4.10)$$

For this problem, the semidefinite relaxation (3.4.7) after evident simplifications becomes the semidefinite program

$$\begin{aligned} & \text{maximize} && \frac{1}{4} \sum_{i,j=1}^n a_{ij}(1 - X_{ij}) \\ & \text{s.t.} && X = [X_{ij}]_{i,j=1}^n = X^T \succeq 0 \\ & && X_{ii} = 1, i = 1, \dots, n; \end{aligned} \quad (3.4.11)$$

the optimal value in the latter problem is an upper bound for the optimal value of MAXCUT.

The fact that (3.4.11) is a relaxation of (3.4.10) can be established directly, independently of any “general theory”: (3.4.10) is the problem of maximizing the objective

$$\frac{1}{4} \sum_{i,j=1}^n a_{ij} - \frac{1}{2} \sum_{i,j=1}^n a_{ij} x_i x_j \equiv \frac{1}{4} \sum_{i,j=1}^n a_{ij} - \frac{1}{4} \text{Tr}(AX(x)), \quad X(x) = xx^T$$

over all rank 1 matrices $X(x) = xx^T$ given by n -dimensional vectors x with entries ± 1 . All these matrices are symmetric positive semidefinite with unit entries on the diagonal, i.e., they belong the feasible set of (3.4.11). Thus, (3.4.11) indeed is a relaxation of (3.4.10).

The quality of the semidefinite relaxation (3.4.11) is given by the following brilliant result of Goemans and Williamson (1995):

Theorem 3.4.1 *Let OPT be the optimal value of the MAXCUT problem (3.4.10), and SDP be the optimal value of the semidefinite relaxation (3.4.11). Then*

$$OPT \leq SDP \leq \alpha \cdot OPT, \quad \alpha = 1.138\dots \quad (3.4.12)$$

Proof. The left inequality in (3.4.12) is what we already know – it simply says that semidefinite program (3.4.11) is a relaxation of MAXCUT. To get the right inequality, Goemans and Williamson act as follows. Let $X = [X_{ij}]$ be a feasible solution to the semidefinite relaxation. Since X is positive semidefinite, it is the covariance matrix of a Gaussian random vector ξ with zero mean, so that $\mathbf{E}\{\xi_i \xi_j\} = X_{ij}$. Now consider the random vector $\zeta = \text{sign}[\xi]$ comprised of signs of the entries in ξ . A realization of ζ is almost surely a vector with coordinates ± 1 , i.e., it is a cut. What is the expected weight of this cut? A straightforward computation demonstrates that $\mathbf{E}\{\zeta_i \zeta_j\} = \frac{2}{\pi} \text{asin}(X_{ij})$ ¹⁴⁾. It follows that

$$\mathbf{E} \left\{ \frac{1}{4} \sum_{i,j=1}^n a_{ij} (1 - \zeta_i \zeta_j) \right\} = \frac{1}{4} \sum_{i,j=1}^n a_{ij} \left(1 - \frac{2}{\pi} \text{asin}(X_{ij}) \right). \quad (3.4.13)$$

Now, it is immediately seen that

$$-1 \leq t \leq 1 \Rightarrow 1 - \frac{2}{\pi} \text{asin}(t) \geq \alpha^{-1}(1 - t), \quad \alpha = 1.138\dots$$

In view of $a_{ij} \geq 0$, the latter observation combines with (3.4.13) to imply that

$$\mathbf{E} \left\{ \frac{1}{4} \sum_{i,j=1}^n a_{ij} (1 - \zeta_i \zeta_j) \right\} \geq \alpha^{-1} \frac{1}{4} \sum_{i,j=1}^n a_{ij} (1 - X_{ij}).$$

The left hand side in this inequality, by evident reasons, is $\leq OPT$. We have proved that the value of the objective in (3.4.11) at every feasible solution X to the problem is $\leq \alpha \cdot OPT$, whence $SDP \leq \alpha \cdot OPT$ as well. ■

Note that the proof of Theorem 3.4.1 provides a randomized algorithm for building a sub-optimal, within the factor $\alpha^{-1} = 0.878\dots$, solution to MAXCUT: we find a (nearly) optimal solution X to the semidefinite relaxation (3.4.11) of MAXCUT, generate a sample of, say, 100 realizations of the associated random cuts ζ and choose the one with the maximum weight.

¹⁴⁾Recall that $X_{ij} \succeq 0$ is normalized by the requirement $X_{ii} = 1$ for all i . Omitting this normalization, we would get $\mathbf{E}\{\zeta_i \zeta_j\} = \frac{2}{\pi} \text{asin}\left(\frac{X_{ij}}{\sqrt{X_{ii} X_{jj}}}\right)$.

3.4.1.6 Nesterov's $\frac{\pi}{2}$ Theorem

In the MAXCUT problem, we are in fact maximizing the homogeneous quadratic form

$$x^T Ax \equiv \sum_{i=1}^n \left(\sum_{j=1}^n a_{ij} \right) x_i^2 - \sum_{i,j=1}^n a_{ij} x_i x_j$$

over the set S_n of n -dimensional vectors x with coordinates ± 1 . It is easily seen that the matrix A of this form is positive semidefinite and possesses a specific feature that the off-diagonal entries are nonpositive, while the sum of the entries in every row is 0. What happens when we are maximizing over S_n a quadratic form $x^T Ax$ with a general-type (symmetric) matrix A ? An extremely nice result in this direction was obtained by Yu. Nesterov. The cornerstone of Nesterov's construction relates to the case when A is positive semidefinite, and this is the case we shall focus on. Note that the problem of maximizing a quadratic form $x^T Ax$ with positive semidefinite (and, say, integer) matrix A over S_n , same as MAXCUT, is NP-complete.

The semidefinite relaxation of the problem

$$\max_x \left\{ x^T Ax : x \in S_n \quad [\Leftrightarrow x_i \in \{-1, 1\}, i = 1, \dots, n] \right\} \quad (3.4.14)$$

can be built exactly in the same way as (3.4.11) and turns out to be the semidefinite program

$$\begin{aligned} & \text{maximize} && \text{Tr}(AX) \\ & \text{s.t.} && \\ & X = X^T = [X_{ij}]_{i,j=1}^n &\succeq 0 \\ & X_{ii} &= 1, \quad i = 1, \dots, n. \end{aligned} \quad (3.4.15)$$

The optimal value in this problem, let it again be called SDP , is \geq the optimal value OPT in the original problem (3.4.14). The ratio OPT/SDP , however, cannot be too large:

Theorem 3.4.2 [Nesterov's $\frac{\pi}{2}$ Theorem] *Let A be positive semidefinite. Then*

$$OPT \leq SDP \leq \frac{\pi}{2} SDP \quad [\frac{\pi}{2} = 1.570\dots]$$

The proof utilizes the central idea of Goemans and Williamson in the following brilliant reasoning:

The inequality $SDP \geq OPT$ is valid since (3.4.15) is a relaxation of (3.4.14). Let X be a feasible solution to the relaxed problem; let, same as in the MAXCUT construction, ξ be a Gaussian random vector with zero mean and the covariance matrix X , and let $\zeta = \text{sign}[\xi]$. As we remember,

$$\mathbf{E} \{ \zeta^T A \zeta \} = \sum_{i,j} A_{ij} \frac{2}{\pi} \text{asin}(X_{ij}) = \frac{2}{\pi} \text{Tr}(A, \text{asin}[X]), \quad (3.4.16)$$

where for a function f on the axis and a matrix X $f[X]$ denotes the matrix with the entries $f(X_{ij})$. Now – the crucial (although simple) observation:

For a positive semidefinite symmetric matrix X with diagonal entries ± 1 (in fact, for any positive semidefinite X with $|X_{ij}| \leq 1$) one has

$$\text{asin}[X] \succeq X. \quad (3.4.17)$$

The proof is immediate: denoting by $[X]^k$ the matrix with the entries X_{ij}^k and making use of the Taylor series for the asin (this series converges uniformly on $[-1, 1]$), for a matrix X with all entries belonging to $[-1, 1]$ we get

$$\text{asin}[X] - X = \sum_{k=1}^{\infty} \frac{1 \times 3 \times 5 \times \dots \times (2k-1)}{2^k k!(2k+1)} [X]^{2k+1},$$

and all we need is to note is that all matrices in the left hand side are $\succeq 0$ along with X ¹⁵⁾

Combining (3.4.16), (3.4.17) and the fact that A is positive semidefinite, we conclude that

$$[OPT \geq] \quad \mathbf{E} \{ \zeta^T A \zeta \} = \frac{2}{\pi} \text{Tr}(A \text{asin}[X]) \geq \frac{2}{\pi} \text{Tr}(AX).$$

The resulting inequality is valid for every feasible solution X of (3.4.15), whence $SDP \leq \frac{\pi}{2} OPT$. ■

The $\frac{\pi}{2}$ Theorem has a number of far-reaching consequences (see Nesterov's papers [27, 28]), for example, the following two:

Theorem 3.4.3 *Let T be an SDr compact subset in \mathbf{R}_+^n . Consider the set*

$$\mathcal{T} = \{x \in \mathbf{R}^n : (x_1^2, \dots, x_n^2)^T \in T\},$$

and let A be a symmetric $n \times n$ matrix. Then the quantities $m_(A) = \min_{x \in \mathcal{T}} x^T Ax$ and $m^*(A) = \max_{x \in \mathcal{T}} x^T Ax$ admit efficiently computable bounds*

$$\begin{aligned} s_*(A) &\equiv \min_X \left\{ \text{Tr}(AX) : X \succeq 0, (X_{11}, \dots, X_{nn})^T \in T \right\}, \\ s^*(A) &\equiv \max_X \left\{ \text{Tr}(AX) : X \succeq 0, (X_{11}, \dots, X_{nn})^T \in T \right\}, \end{aligned}$$

such that

$$s_*(A) \leq m_*(A) \leq m^*(A) \leq s^*(A)$$

and

$$m^*(A) - m_*(A) \leq s^*(A) - s_*(A) \leq \frac{\pi}{4-\pi} (m^*(A) - m_*(A))$$

(in the case of $A \succeq 0$ and $0 \in T$, the factor $\frac{\pi}{4-\pi}$ can be replaced with $\frac{\pi}{2}$).

Thus, the "variation" $\max_{x \in \mathcal{T}} x^T Ax - \min_{x \in \mathcal{T}} x^T Ax$ of the quadratic form $x^T Ax$ on \mathcal{T} can be efficiently bounded from above, and the bound is tight within an absolute constant factor.

Note that if T is given by a strictly feasible SDR, then both $(-s_(A))$ and $s^*(A)$ are SDr functions of A (Proposition 2.4.4).*

Theorem 3.4.4 *Let $p \in [2, \infty]$, $r \in [1, 2]$, and let A be an $m \times n$ matrix. Consider the problem of computing the operator norm $\|A\|_{p,r}$ of the linear mapping $x \mapsto Ax$, considered as the mapping from the space \mathbf{R}^n equipped with the norm $\|\cdot\|_p$ to the space \mathbf{R}^m equipped with the norm $\|\cdot\|_r$:*

$$\|A\|_{p,r} = \max \{ \|Ax\|_r : \|x\|_p \leq 1 \};$$

¹⁵⁾The fact that the entry-wise product of two positive semidefinite matrices is positive semidefinite is a standard fact from Linear Algebra. The easiest way to understand it is to note that if P, Q are positive semidefinite symmetric matrices of the same size, then they are Gram matrices: $P_{ij} = p_i^T p_j$ for certain system of vectors p_i from certain (no matter from which exactly) \mathbf{R}^N and $Q_{ij} = q_i^T q_j$ for a system of vectors q_i from certain \mathbf{R}^M . But then the entry-wise product of P and Q – the matrix with the entries $P_{ij}Q_{ij} = (p_i^T p_j)(q_i^T q_j)$ – also is a Gram matrix, namely, the Gram matrix of the matrices $p_i q_i^T \in \mathbf{M}^{N,M} = \mathbf{R}^{NM}$. Since every Gram matrix is positive semidefinite, the entry-wise product of P and Q is positive semidefinite.

note that it is difficult (*NP-hard*) to compute this norm, except for the case of $p = r = 2$. The “computationally intractable” quantity $\|A\|_{p,r}$ admits an efficiently computable upper bound

$$\omega_{p,r}(A) = \min_{\lambda \in \mathbf{R}^m, \mu \in \mathbf{R}^n} \left\{ \frac{1}{2} \left[\|\mu\|_{\frac{p}{p-2}} + \|\lambda\|_{\frac{r}{2-r}} \right] : \begin{pmatrix} \text{Diag}\{\mu\} & A^T \\ A & \text{Diag}\{\lambda\} \end{pmatrix} \succeq 0 \right\};$$

this bound is exact for a nonnegative matrix A , and for an arbitrary A the bound is tight within the factor $\frac{\pi}{2\sqrt{3}-2\pi/3} = 2.293\dots$:

$$\|A\|_{p,r} \leq \omega_{p,r}(A) \leq \frac{\pi}{2\sqrt{3}-2\pi/3} \|A\|_{p,r}.$$

Moreover, when $p \in [1, \infty)$ and $r \in [1, 2]$ are rational (or $p = \infty$ and $r \in [1, 2]$ is rational), the bound $\omega_{p,r}(A)$ is an SDr function of A .

3.4.2 Matrix Cube Theorem and interval stability analysis/synthesis

Consider the problem of Lyapunov Stability Analysis in the case of interval uncertainty:

$$\mathcal{U} = \mathcal{U}_\rho = \{A \in \mathbf{M}^{n,n} \mid |A_{ij} - A_{ij}^*| \leq \rho D_{ij}, i, j = 1, \dots, n\}, \quad (3.4.18)$$

where A^* is the “nominal” matrix, $D \neq 0$ is a matrix with nonnegative entries specifying the “scale” for perturbations of different entries, and $\rho \geq 0$ is the “level of perturbations”. We deal with a polytopic uncertainty, and as we remember from Section 3.3.3, to certify the stability is the same as to find a feasible solution of the associated semidefinite program (3.3.8) with a negative value of the objective. The difficulty, however, is that the number N of LMI constraints in this problem is the number of vertices of the polytope (3.4.18), i.e., $N = 2^m$, where m is the number of uncertain entries in our interval matrix (\equiv the number of positive entries in D). For 5×5 interval matrices with “full uncertainty” $m = 25$, i.e., $N = 2^{25} = 33,554,432$, which is “a bit” too many; for “fully uncertain” 10×10 matrices, $N = 2^{100} > 1.2 \times 10^{30}\dots$ Thus, the “brute force” approach fails already for “pretty small” matrices affected by interval uncertainty.

In fact, the difficulty we have encountered lies in the NP-hardness of the following problem:

Given a candidate Lyapunov stability certificate $X \succ 0$ and $\rho > 0$, check whether X indeed certifies stability of all instances of \mathcal{U}_ρ , i.e., whether X solves the semi-infinite system of LMI’s

$$A^T X + X A \preceq -I \quad \forall A \in \mathcal{U}_\rho. \quad (3.4.19)$$

(in fact, we are interested in the system $A^T X + X A \prec 0 \forall A \in \mathcal{U}_\rho$, but this is a minor difference – the “system of interest” is homogeneous in X , and therefore every feasible solution of it can be converted to a solution of (3.4.19) just by scaling $X \mapsto tX$).

The above problem, in turn, is a particular case of the following problem:

“Matrix Cube”: Given matrices $A_0, A_1, \dots, A_m \in \mathbf{S}^n$ with $A_0 \succeq 0$, find the largest $\rho = R[A_1, \dots, A_m : A_0]$ such that the set

$$\mathcal{A}_\rho = \left\{ A = A_0 + \sum_{i=1}^m z_i A_i \mid \|z\|_\infty \leq \rho \right\} \quad (3.4.20)$$

– the image of the m -dimensional cube $\{z \in \mathbf{R}^m \mid \|z\|_\infty \leq \rho\}$ under the affine mapping $z \mapsto A_0 + \sum_{i=1}^m z_i A_i$ – is contained in the semidefinite cone \mathbf{S}_+^n .

This is the problem we will focus on.

The Matrix Cube Theorem. The problem ‘‘Matrix Cube’’ (MC for short) is NP-hard; this is true also for the ‘‘feasibility version’’ MC_ρ of MC, where we, given a $\rho \geq 0$, are interested to verify the inclusion $\mathcal{A}_\rho \subset \mathbf{S}_+^n$. However, we can point out a simple *sufficient condition* for the validity of the inclusion $\mathcal{A}_\rho \subset \mathbf{S}_+^n$:

Proposition 3.4.4 *Assume that the system of LMI’s*

$$\begin{aligned} (a) \quad & X^i \succeq \rho A_i, \quad X^i \succeq -\rho A_i, \quad i = 1, \dots, m; \\ (b) \quad & \sum_{i=1}^m X_i \preceq A_0 \end{aligned} \tag{S}_\rho$$

in matrix variables $X^1, \dots, X^m \in \mathbf{S}^n$ is solvable. Then $\mathcal{A}_\rho \subset \mathbf{S}_+^n$.

Proof. Let X^1, \dots, X^m be a solution of (S_ρ) . From (a) it follows that whenever $\|z\|_\infty \leq \rho$, we have $X^i \succeq z_i A_i$ for all i , whence by (b)

$$A_0 + \sum_{i=1}^m z_i A_i \succeq A_0 - \sum_i X_i \succeq 0. \quad \blacksquare$$

Our main result is that the sufficient condition for the inclusion $\mathcal{A}_\rho \subset \mathbf{S}_+^n$ stated by Proposition 3.4.4 is not too conservative:

Theorem 3.4.5 *If the system of LMI’s (S_ρ) is not solvable, then*

$$\mathcal{A}_{\vartheta(\mu)\rho} \not\subset \mathbf{S}_+^n; \tag{3.4.21}$$

here

$$\mu = \max_{1 \leq i \leq m} \text{Rank}(A_i)$$

(note ‘‘ $i \geq 1$ ’’ in the max!), and

$$\vartheta(k) \leq \frac{\pi\sqrt{k}}{2}, \quad k \geq 1; \quad \vartheta(2) = \frac{\pi}{2}. \tag{3.4.22}$$

Proof. Below $\zeta \sim \mathcal{N}(0, I_n)$ means that ζ is a random Gaussian n -dimensional vector with zero mean and the unit covariance matrix, and $p_n(\cdot)$ stands for the density of the corresponding probability distribution:

$$p_n(u) = (2\pi)^{-n/2} \exp \left\{ -\frac{u^T u}{2} \right\}, \quad u \in \mathbf{R}^n.$$

Let us set

$$\vartheta(k) = \frac{1}{\min \left\{ \int |\alpha_i u_1^2 + \dots + \alpha_k u_k^2| p_k(u) du \mid \alpha \in \mathbf{R}^k, \|\alpha\|_1 = 1 \right\}}. \tag{3.4.23}$$

It suffices to verify that

- (i): With the just defined $\vartheta(\cdot)$, insolvability of (S_ρ) does imply (3.4.21);
- (ii): $\vartheta(\cdot)$ satisfies (3.4.22).

Let us prove (i).

1⁰. Assume that (S_ρ) has no solutions. It means that the optimal value of the semidefinite problem

$$\min_{t, \{X^i\}} \left\{ t \mid \begin{array}{l} X^i \succeq \rho A_i, \quad X^i \succeq -\rho A_i, \quad i = 1, \dots, m; \\ \sum_{i=1}^m X_i \preceq A_0 + tI \end{array} \right\} \tag{3.4.24}$$

is positive. Since the problem is strictly feasible, its optimal value is positive if and only if the optimal value of the dual problem

$$\max_{W, \{U^i, V^i\}} \left\{ \rho \sum_{i=1}^m \text{Tr}([U^i - V^i]A_i) - \text{Tr}(WA_0) \middle| \begin{array}{l} U^i + V^i = W, i = 1, \dots, m, \\ \text{Tr}(W) = 1, \\ U^i, V^i, W \succeq 0 \end{array} \right\}$$

is positive. Thus, there exists matrices U^i, V^i, W such that

$$\begin{aligned} (a) \quad & U^i, V^i, W \succeq 0, \\ (b) \quad & U^i + V^i = W, i = 1, 2, \dots, m, \\ (c) \quad & \rho \sum_{i=1}^m \text{Tr}([U^i - V^i]A_i) > \text{Tr}(WA_0). \end{aligned} \tag{3.4.25}$$

²⁰. Now let us use simple

Lemma 3.4.1 *Let $W, A \in \mathbf{S}^n$, $W \succeq 0$. Then*

$$\max_{U, V \succeq 0, U+V=W} \text{Tr}([U - V]A) = \max_{X=X^T: \|\lambda(X)\|_\infty \leq 1} \text{Tr}(XW^{1/2}AW^{1/2}) = \|\lambda(W^{1/2}AW^{1/2})\|_1. \tag{3.4.26}$$

Proof of Lemma. We clearly have

$$U, V \succeq 0, U + V = W \Leftrightarrow U = W^{1/2}PW^{1/2}, V = W^{1/2}QW^{1/2}, P, Q \succeq 0, P + Q = I,$$

whence

$$\max_{U, V: U, V \succeq 0, U+V=W} \text{Tr}([U - V]A) = \max_{P, Q: P, Q \succeq 0, P+Q=I} \text{Tr}([P - Q]W^{1/2}AW^{1/2}).$$

When P, Q are linked by the relation $P + Q = I$ and vary in $\{P \succeq 0, Q \succeq 0\}$, the matrix $X = P - Q$ runs through the entire “interval” $\{-I \preceq X \preceq I\}$ (why?); we have proved the first equality in (3.4.26). When proving the second equality, we may assume w.l.o.g. that the matrix $W^{1/2}AW^{1/2}$ is diagonal, so that $\text{Tr}(XW^{1/2}AW^{1/2}) = \lambda^T(W^{1/2}AW^{1/2})\text{Dg}(X)$, where $\text{Dg}(X)$ is the diagonal of X . When X runs through the “interval” $\{-I \preceq X \preceq I\}$, the diagonal of X runs through the entire unit cube $\{\|x\|_\infty \leq 1\}$, which immediately yields the second equality in (3.4.26). ■

By Lemma 3.4.1, from (3.4.25) it follows that there exists $W \succeq 0$ such that

$$\rho \sum_{i=1}^m \|\lambda(W^{1/2}A_iW^{1/2})\|_1 > \text{Tr}(W^{1/2}A_0W^{1/2}). \tag{3.4.27}$$

³⁰. Now let us use the following observation:

Lemma 3.4.2 *With $\xi \sim \mathcal{N}(0, I_n)$, for every k and every symmetric $n \times n$ matrix A with $\text{Rank}(A) \leq k$ one has*

$$\begin{aligned} (a) \quad & \mathbf{E}\{\xi^T A \xi\} = \text{Tr}(A), \\ (a) \quad & \mathbf{E}\{|\xi^T A \xi|\} \geq \frac{1}{\vartheta(\text{Rank}(A))} \|\lambda(A)\|_1; \end{aligned} \tag{3.4.28}$$

here \mathbf{E} stands for the expectation w.r.t. the distribution of ξ .

Proof of Lemma. (3.4.28.a) is evident:

$$\mathbf{E}\{\xi^T A \xi\} = \sum_{i,j=1}^m A_{ij} \mathbf{E}\{\xi_i \xi_j\} = \text{Tr}(A).$$

To prove (3.4.28.b), by homogeneity it suffices to consider the case when $\|\lambda(A)\|_1 = 1$, and by rotational invariance of the distribution of ξ – the case when A is diagonal, and the first $\text{Rank}(A)$ of diagonal entries of A are the nonzero eigenvalues of the matrix; with this normalization, the required relation immediately follows from the definition of $\vartheta(\cdot)$. ■

4^0 . Now we are ready to prove (i). Let $\xi \sim \mathcal{N}(0, I_n)$. We have

$$\begin{aligned} \mathbf{E} \left\{ \rho \vartheta(\mu) \sum_{i=1}^k |\xi^T W^{1/2} A_i W^{1/2} \xi| \right\} &= \sum_{i=1}^m \rho \vartheta(\mu) \mathbf{E} \{ |\xi^T W^{1/2} A_i W^{1/2} \xi| \} \\ &\geq \rho \sum_{i=1}^m \|\lambda(W^{1/2} A_i W^{1/2})\|_1 \\ &\quad [\text{by (3.4.28.b) due to } \text{Rank}(W^{1/2} A_i W^{1/2}) \leq \text{Rank}(A_i) \leq \mu, i \geq 1] \\ &> \text{Tr}(W^{1/2} A_0 W^{1/2}) \quad [\text{by (3.4.27)}] \\ &= \text{Tr}(\xi^T W^{1/2} A_0 W^{1/2} \xi), \quad [\text{by (3.4.28.a)}] \end{aligned}$$

whence

$$\mathbf{E} \left\{ \rho \vartheta(\mu) \sum_{i=1}^k |\xi^T W^{1/2} A_i W^{1/2} \xi| - \xi^T W^{1/2} A_0 W^{1/2} \xi \right\} > 0.$$

It follows that there exists $r \in \mathbf{R}^n$ such that

$$\vartheta(\mu) \rho \sum_{i=1}^m |r^T W^{1/2} A_i W^{1/2} r| > r^T W^{1/2} A_0 W^{1/2} r,$$

so that setting $z_i = -\vartheta(\mu) \rho \text{sign}(r^T W^{1/2} A_i W^{1/2} r)$, we get

$$r^T W^{1/2} \left(A_0 + \sum_{i=1}^m z_i A_i \right) W^{1/2} r < 0.$$

We see that the matrix $A_0 + \sum_{i=1}^m z_i A_i$ is not positive semidefinite, while by construction $\|z\|_\infty \leq \vartheta(\mu) \rho$.

Thus, (3.4.21) holds true. (i) is proved.

To prove (ii), let $\alpha \in \mathbf{R}^k$ be such that $\|\alpha\|_1 = 1$, and let

$$J = \int |\alpha_1 u_1^2 + \dots + \alpha_k u_k^2| p_k(u) du.$$

Let $\beta = \begin{bmatrix} \alpha \\ -\alpha \end{bmatrix}$, and let $\xi \sim \mathcal{N}(0, I_{2k})$. We have

$$\mathbf{E} \left\{ \left| \sum_{i=1}^{2k} \beta_i \xi_i^2 \right| \right\} \leq \mathbf{E} \left\{ \left| \sum_{i=1}^k \beta_i \xi_i^2 \right| \right\} + \mathbf{E} \left\{ \left| \sum_{i=1}^k \beta_{i+k} \xi_{i+k}^2 \right| \right\} = 2J. \quad (3.4.29)$$

On the other hand, let $\eta_i = \frac{1}{\sqrt{2}}(\xi_i - \xi_{k+i})$, $\zeta_i = \frac{1}{\sqrt{2}}(\xi_i + \xi_{k+i})$, $i = 1, \dots, k$, and let $\omega = \begin{pmatrix} \alpha_1 \eta_1 \\ \vdots \\ \alpha_k \eta_k \end{pmatrix}$,

$\tilde{\omega} = \begin{pmatrix} |\alpha_1 \eta_1| \\ \vdots \\ |\alpha_k \eta_k| \end{pmatrix}$, $\zeta = \begin{pmatrix} \zeta_1 \\ \vdots \\ \zeta_k \end{pmatrix}$. Observe that ζ and ω are independent and $\zeta \sim \mathcal{N}(0, I_k)$. We have

$$\mathbf{E} \left\{ \left| \sum_{i=1}^{2k} \beta_i \xi_i^2 \right| \right\} = 2 \mathbf{E} \left\{ \left| \sum_{i=1}^k \alpha_i \eta_i \zeta_i \right| \right\} = 2 \mathbf{E} \{ |\omega^T \zeta| \} = \mathbf{E} \{ \|\omega\|_2 \} \mathbf{E} \{ |\zeta_1| \},$$

where the concluding equality follows from the fact that $\zeta \sim \mathcal{N}(0, I_k)$ is independent of ω . We further have

$$\mathbf{E} \{ |\zeta_1| \} = \int |t| p_1(t) dt = \frac{2}{\sqrt{2\pi}}$$

and

$$\mathbf{E} \{\|\omega\|_2\} = \mathbf{E} \{\|\tilde{\omega}\|_2\} \geq \|\mathbf{E} \{\tilde{\omega}\}\|_2 = \left[\int |t| p_1(t) dt \right] \sqrt{\sum_{i=1}^m \alpha_i^2}.$$

Combining our observations, we come to

$$\mathbf{E} \left\{ \left| \sum_{i=1}^{2k} \beta_i \xi_i^2 \right| \right\} \geq 2 \left(\frac{2}{\sqrt{2\pi}} \right)^2 \|\alpha\|_2 \geq \frac{4}{\pi\sqrt{k}} \|\alpha\|_1 = \frac{4}{\pi\sqrt{k}}.$$

This relation combines with (3.4.29) to yield $J \geq \frac{2}{\pi\sqrt{k}}$. Recalling the definition of $\vartheta(k)$, we come to $\vartheta(k) \leq \frac{\pi\sqrt{k}}{2}$, as required in (3.4.22).

It remains to prove that $\vartheta(2) = \frac{\pi}{2}$. From the definition of $\vartheta(\cdot)$ it follows that

$$\vartheta^{-1}(2) = \min_{0 \leq \theta \leq 1} \int |\theta u_1^2 - (1-\theta)u_2^2| p_2(u) du \equiv \min_{0 \leq \theta \leq 1} f(\theta).$$

The function $f(\theta)$ is clearly convex and satisfies the identity $f(\theta) = f(1-\theta)$, $0 \leq \theta \leq 1$, so that its minimum is attained at $\theta = \frac{1}{2}$. A direct computation says that $f(\frac{1}{2}) = \frac{2}{\pi}$. ■

Corollary 3.4.1 *Let the ranks of all matrices A_1, \dots, A_m in MC be $\leq \mu$. Then the optimal value in the semidefinite problem*

$$\rho[A_1, \dots, A_m : A_0] = \max_{\rho, X^i} \left\{ \rho \mid \begin{array}{l} X^i \succeq \rho A_i, \quad X^i \succeq -\rho A_i, \quad i = 1, \dots, m, \\ \sum_{i=1}^m X^i \preceq A_0 \end{array} \right\} \quad (3.4.30)$$

is a lower bound on $R[A_1, \dots, A_m : A_0]$, and the “true” quantity is at most $\vartheta(\mu)$ times (see (3.4.23), (3.4.22)) larger than the bound:

$$\rho[A_1, \dots, A_m : A_0] \leq R[A_1, \dots, A_m : A_0] \leq \vartheta(\mu) \rho[A_1, \dots, A_m : A_0]. \quad (3.4.31)$$

Application: Lyapunov Stability Analysis for an interval matrix. Now we are equipped to attack the problem of certifying the stability of uncertain linear dynamic system with interval uncertainty. The problem we are interested in is as follows:

“Interval Lyapunov”: Given a stable $n \times n$ matrix A_* ¹⁶⁾ and an $n \times n$ matrix $D \neq 0$ with nonnegative entries, find the supremum $R[A_*, D]$ of those $\rho \geq 0$ for which all instances of the “interval matrix”

$$\mathcal{U}_\rho = \{A \in \mathbf{M}^{n,n} : |A_{ij} - (A_*)_{ij}| \leq \rho D_{ij}, \quad i, j = 1, \dots, n\}$$

share a common quadratic Lyapunov function, i.e., the semi-infinite system of LMI’s

$$X \succeq I; \quad A^T X + X A \preceq -I \quad \forall A \in \mathcal{U}_\rho \quad (\text{Ly}_\rho)$$

in matrix variable $X \in \mathbf{S}^n$ is solvable.

¹⁶⁾I.e., with all eigenvalues from the open left half-plane, or, which is the same, such that $A_*^T X + X A_* \prec 0$ for certain $X \succ 0$.

Observe that $X \succeq I$ solves (Ly_ρ) if and only if the matrix cube

$$\begin{aligned}\mathcal{A}_\rho[X] &= \left\{ B = \underbrace{\left[-I - A_*^T X - X A_* \right]}_{A_0[X]} + \sum_{(i,j) \in \mathcal{D}} z_{ij} \underbrace{\left[[D_{ij} E^{ij}]^T X + X [D_{ij} E^{ij}] \right]}_{A_{ij}[X]} \mid |z_{ij}| \leq \rho, (i,j) \in \mathcal{D} \right\} \\ \mathcal{D} &= \{(i,j) : D_{ij} > 0\}\end{aligned}$$

is contained in \mathbf{S}_+^n ; here E^{ij} are the “basic $n \times n$ matrices” (i,j -th entry of E^{ij} is 1, all other entries are zero). Note that the ranks of the matrices $A_{ij}[X]$, $(i,j) \in \mathcal{D}$, are at most 2. Therefore from Proposition 3.4.4 and Theorem 3.4.5 we get the following result:

Proposition 3.4.5 *Let $\rho \geq 0$. Then*

(i) *If the system of LMI's*

$$\begin{aligned}X \succeq I, \\ X^{ij} \succeq -\rho D_{ij} \left[[E^{ij}]^T X + X E^{ij} \right], \quad X^{ij} \succeq \rho D_{ij} \left[[E^{ij}]^T X + X E^{ij} \right], \quad (i,j) \in \mathcal{D} \\ \sum_{(i,j) \in \mathcal{D}} X^{ij} \preceq -I - A_*^T X - X A_*\end{aligned}\tag{A_\rho}$$

in matrix variables $X, X^{ij}, (i,j) \in \mathcal{D}$, is solvable, then so is the system (Ly_ρ) , and the X -component of a solution of the former system solves the latter system.

(ii) *If the system of LMI's (A_ρ) is not solvable, then so is the system $(\text{Ly}_{\frac{\pi}{2}\rho})$.*

In particular, the supremum $\rho[A_, D]$ of those ρ for which (A_ρ) is solvable is a lower bound for $R[A_*, D]$, and the “true” quantity is at most $\frac{\pi}{2}$ times larger than the bound:*

$$\rho[A_*, D] \leq R[A_*, D] \leq \frac{\pi}{2} \rho[A_*, D].$$

Computing $\rho[A_*, D]$. The quantity $\rho[A_*, D]$, in contrast to $R[A_*, D]$, is “efficiently computable”: applying dichotomy in ρ , we can find a high-accuracy approximation of $\rho[A_*, D]$ via solving a small series of semidefinite feasibility problems (A_ρ) . Note, however, that problem (A_ρ) , although “computationally tractable”, is not that simple: in the case of “full uncertainty” ($D_{ij} > 0$ for all i, j) it has $n^2 + n$ matrix variables of the size $n \times n$ each. It turns out that applying semidefinite duality, one can reduce dramatically the sizes of the problem specifying $\rho[A_*, D]$. The resulting (equivalent!) description of the bound is:

$$\frac{1}{\rho[A_*, D]} = \inf_{\lambda, Y, X, \{\eta_i\}} \left\{ \lambda \mid \begin{array}{c} X \succeq I, \\ \left(\begin{array}{cc} Y - \sum_{\ell=1}^m \eta_\ell e_{j_\ell} e_{j_\ell}^T & [X e_{i_1}; X e_{i_2}; \dots; X e_{i_m}] \\ [X e_{i_1}; X e_{i_2}; \dots; X e_{i_m}]^T & \text{Diag}(\eta_1, \dots, \eta_m) \end{array} \right) \succeq 0, \\ A_0[X] \equiv -I - A_*^T X - X A_* \succ 0, \\ Y \preceq \lambda A_0[X] \end{array} \right\}, \tag{3.4.32}$$

where $(i_1, j_1), \dots, (i_m, j_m)$ are the positions of the uncertain entries in our uncertain matrix (i.e., the pairs (i, j) such that $D_{ij} > 0$) and e_1, \dots, e_n are the standard basic orths in \mathbf{R}^n .

Note that the optimization program in (3.4.32) has just two symmetric matrix variables X, Y , a single scalar variable λ and $m \leq n^2$ scalar variables η_i , i.e., totally at most $2n^2 + n + 2$ scalar design variables, which, for large m , is much less than the design dimension of (A_ρ) .

Remark 3.4.1 Note that our results on the Matrix Cube problem can be applied to the interval version of the Lyapunov Stability *Synthesis* problem, where we are interested to find the supremum R of those ρ for which an uncertain controllable system

$$\frac{d}{dt}x(t) = A(t)x(t) + B(t)u(t)$$

with interval uncertainty

$$(A(t), B(t)) \in \mathcal{U}_\rho = \{(A, B) : |A_{ij} - (A_*)_ij| \leq \rho D_{ij}, |B_{i\ell} - (B_*)_{i\ell}| \leq \rho C_{i\ell} \quad \forall i, j, \ell\}$$

admits a linear feedback

$$u(t) = Kx(t)$$

such that all instances $A(t) + B(t)K$ of the resulting closed loop system share a common quadratic Lyapunov function. Here our constructions should be applied to the semi-infinite system of LMI's

$$Y \succeq I, \quad BL + AY + L^T B^T + YA^T \preceq -I \quad \forall (A, B) \in \mathcal{U}_\rho$$

in variables L, Y (see Proposition 3.3.4), and them yield an efficiently computable lower bound on R which is at most $\frac{\pi}{2}$ times less than R .

We have seen that the Matrix Cube Theorem allows to build tight computationally tractable approximations to semi-infinite systems of LMI's responsible for stability of uncertain linear dynamical systems affected by interval uncertainty. The same is true for many other semi-infinite systems of LMI's arising in Control in the presence of interval uncertainty, since in a typical Control-related LMI, a perturbation of a single entry in the underlying data results in a small-rank perturbation of the LMI – a situation well-suited for applying the Matrix Cube Theorem.

Nesterov's Theorem revisited. Our results on the Matrix Cube problem give an alternative proof of Nesterov's $\frac{\pi}{2}$ Theorem (Theorem 3.4.2). Recall that in this theorem we are comparing the true maximum

$$OPT = \max_d \{d^T Ad \mid \|d\|_\infty \leq 1\}$$

of a *positive semidefinite* ($A \succeq 0$) quadratic form on the unit n -dimensional cube and the semidefinite upper bound

$$SDP = \max_X \{\text{Tr}(AX) \mid X \succeq 0, X_{ii} \leq 1, i = 1, \dots, n\} \tag{3.4.33}$$

on OPT ; the theorem says that

$$OPT \leq SDP \leq \frac{\pi}{2} OPT. \tag{3.4.34}$$

To derive (3.4.34) from the Matrix Cube-related considerations, assume that $A \succ 0$ rather than $A \succeq 0$ (by continuity reasons, to prove (3.4.34) for the case of $A \succ 0$ is the same as to prove the relation for all $A \succeq 0$) and let us start with the following simple observation:

Lemma 3.4.3 *Let $A \succ 0$ and*

$$OPT = \max_d \left\{ d^T Ad \mid \|d\|_\infty \leq 1 \right\}.$$

Then

$$\frac{1}{OPT} = \max \left\{ \rho : \begin{pmatrix} 1 & d^T \\ d & A^{-1} \end{pmatrix} \succeq 0 \quad \forall (d : \|d\|_\infty \leq \rho^{1/2}) \right\} \quad (3.4.35)$$

and

$$\frac{1}{OPT} = \max \left\{ \rho : A^{-1} \succeq X \quad \forall (X \in \mathbf{S}^n : |X_{ij}| \leq \rho \forall i, j) \right\}. \quad (3.4.36)$$

Proof. To get (3.4.35), note that by the Schur Complement Lemma, all matrices of the form $\begin{pmatrix} 1 & d^T \\ d & A^{-1} \end{pmatrix}$ with $\|d\|_\infty \leq \rho^{1/2}$ are $\succeq 0$ if and only if $d^T(A^{-1})^{-1}d = d^TAd \leq 1$ for all d , $\|d\|_\infty \leq \rho^{1/2}$, i.e., if and only if $\rho \cdot OPT \leq 1$; we have derived (3.4.35). We now have

$$\begin{aligned} (a) \quad & \frac{1}{OPT} \geq \rho \\ & \Downarrow \\ & \begin{pmatrix} 1 & d^T \\ d & A^{-1} \end{pmatrix} \succeq 0 \quad \forall (d : \|d\|_\infty \leq \rho^{1/2}) \\ & \Downarrow \quad \text{[the Schur Complement Lemma]} \\ & A^{-1} \succeq \rho dd^T \quad \forall (d, \|d\|_\infty \leq 1) \\ & \Downarrow \\ & x^T A^{-1} x \geq \rho(d^T x)^2 \quad \forall x \quad \forall (d : \|d\|_\infty \leq 1) \\ & \Downarrow \\ & x^T A^{-1} x \geq \rho \|x\|_1^2 \quad \forall x \\ & \Downarrow \\ (b) \quad & A^{-1} \succeq \rho Y \quad \forall (Y = Y^T : |Y_{ij}| \leq 1 \forall i, j) \end{aligned}$$

where the concluding \Downarrow is given by the evident relation

$$\|x\|_1^2 = \max_Y \left\{ x^T Y x : Y = Y^T, |Y_{ij}| \leq 1 \forall i, j \right\}.$$

The equivalence $(a) \Leftrightarrow (b)$ is exactly (3.4.36). ■

By (3.4.36), $\frac{1}{OPT}$ is exactly the maximum R of those ρ for which the matrix cube

$$\mathcal{C}_\rho = \left\{ A^{-1} + \sum_{1 \leq i \leq j \leq n} z_{ij} S^{ij} \mid \max_{i,j} |z_{ij}| \leq \rho \right\}$$

is contained in \mathbf{S}_+^n ; here S^{ij} are the “basic symmetric matrices” (S^{ii} has a single nonzero entry, equal to 1, in the cell ii , and S^{ij} , $i < j$, has exactly two nonzero entries, equal to 1, in the cells ij and ji). Since the ranks of the matrices S^{ij} do not exceed 2, Proposition 3.4.4 and Theorem 3.4.5 say that the optimal value in the semidefinite program

$$\rho(A) = \max_{\rho, X^{ij}} \left\{ \rho \mid \begin{array}{l} X^{ij} \succeq \rho S^{ij}, X^{ij} \succeq -\rho S^{ij}, 1 \leq i \leq j \leq n, \\ \sum_{i \leq j} X^{ij} \preceq A^{-1} \end{array} \right\} \quad (\text{S})$$

is a lower bound for R , and this bound coincides with R up to the factor $\frac{\pi}{2}$; consequently, $\frac{1}{\rho(A)}$ is an upper bound on OPT , and this bound is at most $\frac{\pi}{2}$ times larger than OPT . It remains to note that a direct computation demonstrates that $\frac{1}{\rho(A)}$ is exactly the quantity SDP given by (3.4.33).

3.4.3 Robust Quadratic Programming

The concept of robust counterpart of an optimization problem with uncertain data (see Section 2.4.1) is in no sense restricted to Linear Programming. Whenever we have an optimization problem depending on certain data, we may ask what happens when the data are uncertain and all we know is an uncertainty set the data belong to. Given such an uncertainty set, we may require from candidate solutions to be *robust feasible* – to satisfy the realizations of the constraints for all data running through the uncertainty set. The robust counterpart of an uncertain problem is the problem of minimizing the objective¹⁷⁾ over the set of robust feasible solutions.

Now, we have seen in Section 2.4.1 that the “robust form” of an uncertain linear inequality with the coefficients varying in an ellipsoid is a conic quadratic inequality; as a result, the robust counterpart of an uncertain LP problem with ellipsoidal uncertainty (or, more general, with a CQR uncertainty set) is a conic quadratic problem. What is the “robust form” of an uncertain *conic quadratic* inequality

$$\|Ax + b\|_2 \leq c^T x + d \quad [A \in \mathbf{M}^{m,n}, b \in \mathbf{R}^m, c \in \mathbf{R}^n, d \in \mathbf{R}] \quad (3.4.37)$$

with uncertain data $(A, b, c, d) \in \mathcal{U}$? The question is how to describe the set of all robust feasible solutions of this inequality, i.e., the set of x ’s such that

$$\|Ax + b\|_2 \leq c^T x + d \quad \forall (A, b, c, d) \in \mathcal{U}. \quad (3.4.38)$$

We intend to focus on the case when the uncertainty is “side-wise” – the data (A, b) of the left hand side and the data (c, d) of the right hand side of the inequality (3.4.37) *independently of each other* run through respective uncertainty sets $\mathcal{U}_\rho^{\text{left}}, \mathcal{U}^{\text{right}}$ (ρ is the left hand side uncertainty level). It suffices to assume the right hand side uncertainty set to be SDR with a strictly feasible SDR:

$$\mathcal{U}^{\text{right}} = \{(c, d) \mid \exists u : \mathcal{P}c + Qd + Ru \succeq S\}. \quad (3.4.39)$$

As about the left hand side uncertainty set, we assume that it is an intersection of concentric ellipsoids, specifically, that

$$\mathcal{U}_\rho^{\text{left}} = \left\{ [A, b] = [A_*, b_*] + \sum_{\ell=1}^L \zeta_\ell [A_\ell, b_\ell] : \zeta^T Q_j \zeta \leq \rho^2, j = 1, \dots, J \right\}, \quad (3.4.40)$$

where Q_1, \dots, Q_J are positive semidefinite matrices with positive definite sum.

Since the left hand side and the right hand side data independently of each other run through respective uncertainty sets, a point x is robust feasible if and only if there exists a real τ such that

$$\begin{aligned} (a) \quad \tau &\leq c^T x + d \quad \forall (c, d) \in \mathcal{U}^{\text{right}}, \\ (b) \quad \|Ax + b\|_2 &\leq \tau \quad \forall [A, b] \in \mathcal{U}_\rho^{\text{left}}. \end{aligned} \quad (3.4.41)$$

¹⁷⁾Without loss of generality, we may assume that the objective is “certain” – is not affected by the data uncertainty. Indeed, we can always ensure this situation by passing to an equivalent problem with linear (and standard) objective:

$$\min_x \{f(x) : x \in X\} \mapsto \min_{t,x} \{t : f(x) - t \leq 0, x \in X\}.$$

We know from the previous Lecture that the set of (τ, x) satisfying (3.4.41.a) is SDR (see Proposition 2.4.2 and Remark 2.4.1); it is easy to verify that the corresponding SDR is as follows:

$$\begin{array}{ll} (a) & (x, \tau) \text{ satisfies (3.4.41.a)} \\ & \Updownarrow \\ & \exists \Lambda : \\ (b) & \Lambda \succeq 0, \mathcal{P}^* \Lambda = x, \text{Tr}(Q \Lambda) = 1, \mathcal{R}^* \Lambda = 0, \text{Tr}(S \Lambda) \geq \tau. \end{array} \quad (3.4.42)$$

As about building SDR of the set of pairs (τ, x) satisfying (3.4.41.b), this is much more difficult (and in many cases even hopeless) task, since (3.4.38) in general turns out to be NP-hard and as such cannot be posed as an explicit semidefinite program. We can, however, build a kind of “inner approximation” of the set in question. To this end we shall use the ideas of semidefinite relaxation. Specifically, let us set

$$a[x] = A_* x + b_*, \quad A[x] = [A_1 x + b_1, \dots, A_L x + b_L],$$

so that

$$(A_* + \sum_{\ell=1}^L \zeta_\ell A_\ell)x + (b_* + \sum_{\ell=1}^L \zeta_\ell b_\ell) = a[x] + A[x]\zeta.$$

In view of the latter identity, relation (3.4.41.b) reads

$$\|a[x] + \rho A[x]\zeta\|_2 \leq \tau \quad \forall(\zeta : \zeta^T Q_j \zeta \leq 1, j = 1, \dots, J),$$

or, which is the same (set $\zeta = t^{-1}\xi$), as

$$\|ta[x] + \rho A[x]\xi\|_2 \leq \tau t^2 \quad \forall((t, \xi) : \xi^T Q_j \xi \leq t^2, j = 1, \dots, J),$$

which in turn is equivalent to

$$\{\tau \geq 0\}_{\text{I}} \& \left\{ \begin{array}{l} t^2(\tau^2 - a^T[x]a[x]) - 2t\rho a^T[x]A[x]\xi - \rho^2 \xi^T A^T[x]A[x]\xi \geq 0 \\ \forall((\xi, t) : \xi^T Q_j \xi \leq t^2, j = 1, \dots, J) \end{array} \right\}_{\text{II}}.$$

Predicate $\{\cdot\}_{\text{II}}$ requires from certain quadratic form of t, ξ to be nonnegative when a number of other quadratic forms of these variables are nonnegative. An evident sufficient condition for this is that the former quadratic form is \succeq a linear combination, with nonnegative coefficients, of the latter forms. When $\tau \geq 0$, this sufficient condition for the predicate $\{\cdot\}_{\text{II}}$ to be valid can be reduced to the existence of nonnegative weights λ_j such that the quadratic form

$$t^2(\tau^2 - a^T[x]a[x]) - 2t\rho a^T[x]A[x]\xi - \rho^2 \xi^T A^T[x]A[x]\xi - \tau \sum_j \lambda_j(t^2 - \xi^T Q_j \xi)$$

in variables t, ξ is positive semidefinite. This condition is the same as the existence of nonnegative λ_j such that

$$\tau \begin{pmatrix} \tau - \sum_j \lambda_j & \\ & \sum_j \lambda_j Q_j \end{pmatrix} - [a[x], \rho A[x]]^T [a[x], \rho A[x]] \succeq 0.$$

Invoking the Schur Complement Lemma, the latter condition, in turn, is equivalent to the existence of nonnegative λ_j such that the matrix $\begin{pmatrix} \tau - \sum_j \lambda_j & a^T[x] \\ & \sum_j \lambda_j Q_j & \rho A^T[x] \\ a[x] & \rho A[x] & \tau I \end{pmatrix}$ is positive

semidefinite. We have established the implication as follows:

$$(a) \quad \{\tau \succeq 0\} \text{ } \& \left\{ \exists(\lambda_j \geq 0) : \begin{pmatrix} \tau - \sum_j \lambda_j & a^T[x] \\ \sum_j \lambda_j Q_j & \rho A^T[x] \\ \downarrow \\ a[x] & \rho A[x] & \tau I \end{pmatrix} \succeq 0 \right\} \quad (3.4.43)$$

(b) (x, τ) satisfies (3.4.41.b)

Combining our observations, we arrive at the first – easy – part of the following statement:

Proposition 3.4.6 *Let the data in the conic quadratic inequality (3.4.37) be affected by side-wise uncertainty (3.4.39), (3.4.40). Then*

(i) *The system $(S[\rho])$ of LMIs (3.4.42.b), (3.4.43.a) in variables $x, \tau, \Lambda, \{\lambda_j\}$ is a “conservative approximation” of the Robust Counterpart of (3.4.37) in the sense that whenever x can be extended to a feasible solution of $(S[\rho])$, x is robust feasible for (3.4.37), the uncertainty set being $\mathcal{U}_\rho^{\text{left}} \times \mathcal{U}^{\text{right}}$.*

(ii) *The tightness of $(S[\rho])$ as an approximation to the robust counterpart of (3.4.37) can be quantified as follows: if x cannot be extended to a feasible solution of $(S[\rho])$, then x is not robust feasible for (3.4.37), the uncertainty set being $\mathcal{U}_{\vartheta\rho}^{\text{left}} \times \mathcal{U}^{\text{right}}$. Here the “tightness factor” ϑ can be bounded as follows:*

1. *In the case of $J = 1$ (i.e., perturbations ζ are varying in an ellipsoid rather than in an intersection of concentric ellipsoids), one has $\vartheta = 1$ (i.e., $(S[\rho])$ is exactly equivalent to the robust counterpart of $\mathcal{U}_{\vartheta\rho}^{\text{left}} \times \mathcal{U}^{\text{right}}$);*
2. *In the case of “box uncertainty” $J = \dim \zeta$, $\zeta^T Q_j \zeta = \zeta_j^2$, one has $\vartheta = \frac{\pi}{2} = 1.570\dots$;*
3. *In the general case,*

$$\vartheta = \sqrt{2 \ln \left(6 \sum_{j=1}^J \text{Rank}(Q_j) \right)}.$$

For the proof of the “difficult part” (ii) of the Proposition, see [4].

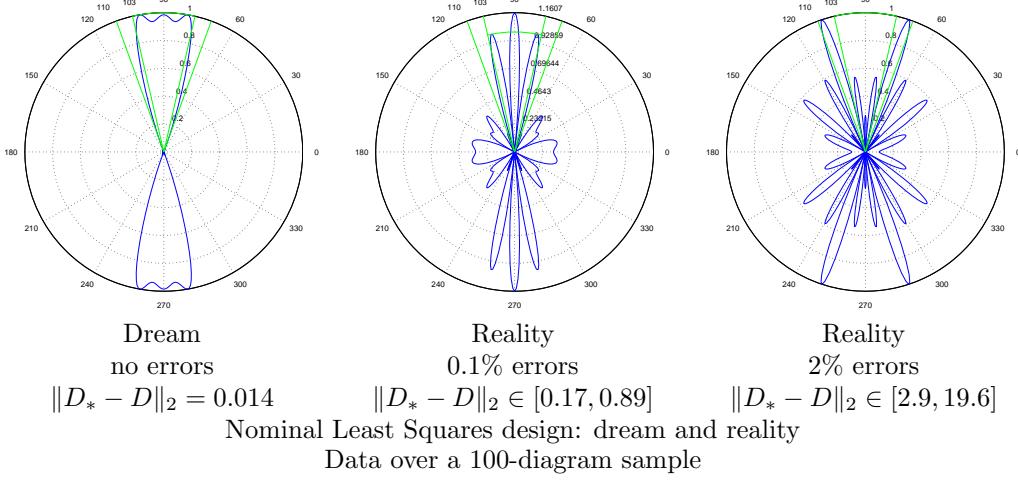
Example: Antenna Synthesis revisited. To illustrate the potential of the Robust Optimization methodology as applied to conic quadratic problems, consider the Circular Antenna Design problem from Section 2.4.1. Assume that now we deal with 40 ring-type antenna elements, and that our goal is to minimize the (discretized) L_2 -distance from the synthesized diagram $\sum_{j=1}^{40} x_j D_{r_{j-1}, r_j}(\cdot)$ to the “ideal” diagram $D_*(\cdot)$ which is equal to 1 in the range $77^\circ \leq \theta \leq 90^\circ$ and is equal to 0 in the range $0^\circ \leq \theta \leq 70^\circ$. The associated problem is just the Least Squares problem

$$\min_{\tau, x} \left\{ \tau : \underbrace{\sqrt{\frac{\sum_{\theta \in \Theta_{\text{cns}}} D_x^2(\theta) + \sum_{\theta \in \Theta_{\text{obj}}} (D_x(\theta) - 1)^2}{\text{card}(\Theta_{\text{cns}} \cup \Theta_{\text{obj}})}}}_{\|D_* - D_x\|_2} \leq \tau \right\}, \quad (3.4.44)$$

$D_x(\theta) = \sum_{j=1}^{40} x_j D_{r_{j-1}, r_j}(\theta)$

where Θ_{cns} and Θ_{obj} are the intersections of the 240-point grid on the segment $0 \leq \theta \leq 90^\circ$ with the “angle of interest” $77^\circ \leq \theta \leq 90^\circ$ and the “sidelobe angle” $0^\circ \leq \theta \leq 70^\circ$, respectively.

The *Nominal Least Squares design* obtained from the optimal solution to this problem is completely unstable w.r.t. small implementation errors $x_j \mapsto (1 + \xi_j)x_j$, $|\xi_j| \leq \rho$:



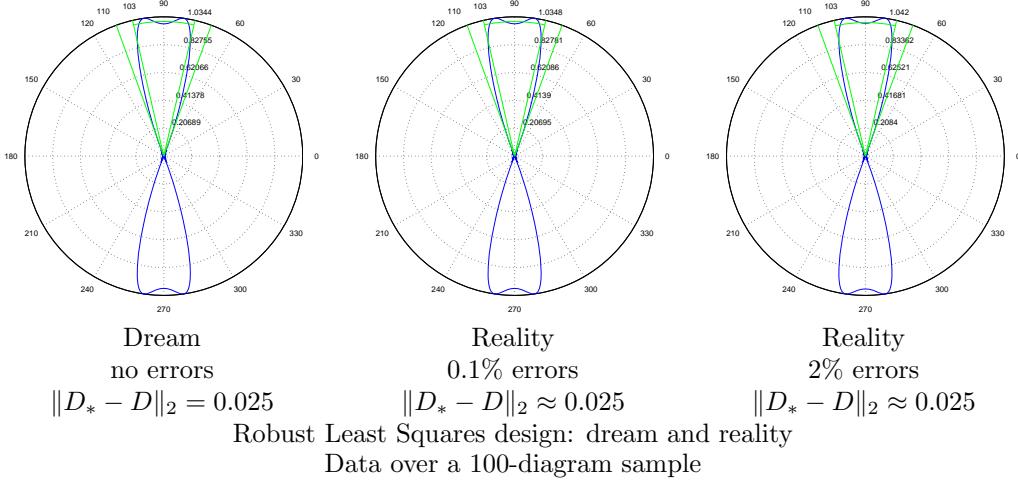
In order to take into account implementation errors, we should treat (3.4.44) as an uncertain conic quadratic problem

$$\left\{ \min_{\tau, x} \{\tau : \|Ax - b\|_2 \leq \tau\} \mid A \in \mathcal{U} \right\}$$

with the uncertainty set of the form

$$\mathcal{U} = \{A = A_* + A_* \text{Diag}(\xi) \mid \|\xi\|_\infty \leq \rho\},$$

which is a particular case of the ellipsoidal uncertainty (specifically, what was called “box uncertainty” in Proposition 3.4.6). In the experiments to be reported, we use $\rho = 0.02$. The approximate Robust Counterpart ($S[\rho]$) of our uncertain conic quadratic problem yields the Robust design as follows:



3.5 *S*-Lemma and Approximate *S*-Lemma

3.5.1 *S*-Lemma

Let us look again at the Lagrange relaxation of a quadratically constrained quadratic problem, but in the very special case when all the forms involved are homogeneous, and the right hand sides of the inequality constraints are zero:

$$\begin{array}{ll} \text{minimize} & x^T Bx \\ \text{s.t.} & x^T A_i x \geq 0, \quad i = 1, \dots, m \end{array} \quad (3.5.1)$$

(B, A_1, \dots, A_m are given symmetric $m \times m$ matrices). Assume that the problem is feasible. In this case (3.5.1) is, at a first glance, a trivial problem: due to homogeneity, its optimal value is either $-\infty$ or 0, depending on whether there exists or does not exist a feasible vector x such that $x^T Bx < 0$. The challenge here is to detect which one of these two alternatives takes place, i.e., to understand whether or not a homogeneous quadratic inequality $x^T Bx \geq 0$ is a consequence of the system of homogeneous quadratic inequalities $x^T A_i x \geq 0$, or, which is the same, to understand when the implication

$$\begin{array}{ll} (a) & x^T A_i x \geq 0, \quad i = 1, \dots, m \\ & \downarrow \\ (b) & x^T Bx \geq 0 \end{array} \quad (3.5.2)$$

holds true.

In the case of homogeneous *linear* inequalities it is easy to recognize when an inequality $x^T b \geq 0$ is a consequence of the system of inequalities $x^T a_i \geq 0, i = 1, \dots, m$: by Farkas Lemma, it is the case if and only if the inequality is a *linear* consequence of the system, i.e., if b is representable as a linear combination, with nonnegative coefficients, of the vectors a_i . Now we are asking a similar question about homogeneous quadratic inequalities: when (b) is a consequence of (a)?

In general, there is no analogy of the Farkas Lemma for homogeneous quadratic inequalities. Note, however, that the easy “if” part of the Lemma can be extended to the quadratic case: if the target inequality (b) can be obtained by linear aggregation of the inequalities (a) and a trivial – identically true – inequality, then the implication in question is true. Indeed, a linear aggregation of the inequalities (a) is an inequality of the type

$$x^T \left(\sum_{i=1}^m \lambda_i A_i \right) x \geq 0$$

with nonnegative weights λ_i , and a trivial – identically true – homogeneous quadratic inequality is of the form

$$x^T Qx \geq 0$$

with $Q \succeq 0$. The fact that (b) can be obtained from (a) and a trivial inequality by linear aggregation means that B can be represented as $B = \sum_{i=1}^m \lambda_i A_i + Q$ with $\lambda_i \geq 0, Q \succeq 0$, or, which is the same, if $B \succeq \sum_{i=1}^m \lambda_i A_i$ for certain nonnegative λ_i . If this is the case, then (3.5.2) is trivially true. We have arrived at the following simple

Proposition 3.5.1 Assume that there exist nonnegative λ_i such that $B \succeq \sum_i \lambda_i A_i$. Then the implication (3.5.2) is true.

Proposition 3.5.1 is no more than a sufficient condition for the implication (3.5.2) to be true, and in general this condition is not necessary. There is, however, an extremely fruitful particular case when the condition is both necessary and sufficient – this is the case of $m = 1$, i.e., a single quadratic inequality in the premise of (3.5.2):

Theorem 3.5.1 [\mathcal{S} -Lemma] Let A, B be symmetric $n \times n$ matrices, and assume that the quadratic inequality

$$x^T Ax \geq 0 \quad (\text{A})$$

is strictly feasible: there exists \bar{x} such that $\bar{x}^T A \bar{x} > 0$. Then the quadratic inequality

$$x^T B x \geq 0 \quad (\text{B})$$

is a consequence of (A) if and only if it is a linear consequence of (A), i.e., if and only if there exists a nonnegative λ such that

$$B \succeq \lambda A.$$

We are about to present an “intelligent” proof of the \mathcal{S} -Lemma based on the ideas of semidefinite relaxation.

In view of Proposition 3.5.1, all we need is to prove the “only if” part of the \mathcal{S} -Lemma, i.e., to demonstrate that if the optimization problem

$$\min_x \left\{ x^T B x : x^T A x \geq 0 \right\}$$

is strictly feasible and its optimal value is ≥ 0 , then $B \succeq \lambda A$ for certain $\lambda \geq 0$. By homogeneity reasons, it suffices to prove exactly the same statement for the optimization problem

$$\min_x \left\{ x^T B x : x^T A x \geq 0, x^T x = n \right\}. \quad (\text{P})$$

The standard semidefinite relaxation of (P) is the problem

$$\min_X \left\{ \text{Tr}(BX) : \text{Tr}(AX) \geq 0, \text{Tr}(X) = n, X \succeq 0 \right\}. \quad (\text{P}')$$

If we could show that when passing from the original problem (P) to the relaxed problem (P') the optimal value (which was nonnegative for (P)) remains nonnegative, we would be done. Indeed, observe that (P') is clearly bounded below (its feasible set is compact!) and is strictly feasible (which is an immediate consequence of the strict feasibility of (A)). Thus, by the Conic Duality Theorem the problem dual to (P') is solvable with the same optimal value (let it be called $n\theta^*$) as the one in (P'). The dual problem is

$$\max_{\mu, \lambda} \left\{ n\mu : \lambda A + \mu I \preceq B, \lambda \geq 0 \right\},$$

and the fact that its optimal value is $n\theta^*$ means that there exists a nonnegative λ such that

$$B \succeq \lambda A + n\theta^* I.$$

If we knew that the optimal value $n\theta^*$ in (P') is nonnegative, we would conclude that $B \succeq \lambda A$ for certain nonnegative λ , which is exactly what we are aiming at. Thus, all we need is to prove that under the premise of the \mathcal{S} -Lemma the optimal value in (P') is nonnegative, and here is the proof:

Observe first that problem (P') is feasible with a compact feasible set, and thus is solvable. Let X^* be an optimal solution to the problem. Since $X^* \geq 0$, there exists a matrix D such that $X^* = DD^T$. Note that we have

$$\begin{aligned} 0 &\leq \text{Tr}(AX^*) = \text{Tr}(ADD^T) = \text{Tr}(D^TAD), \\ n\theta^* &= \text{Tr}(BX^*) = \text{Tr}(BDD^T) = \text{Tr}(D^TBD), \\ n &= \text{Tr}(X^*) = \text{Tr}(DD^T) = \text{Tr}(D^TD). \end{aligned} \tag{*}$$

It remains to use the following observation

(!) *Let P, Q be symmetric matrices such that $\text{Tr}(P) \geq 0$ and $\text{Tr}(Q) < 0$. Then there exists a vector e such that $e^T Pe \geq 0$ and $e^T Qe < 0$.*

Indeed, let us believe that (!) is valid, and let us prove that $\theta^* \geq 0$. Assume, on the contrary, that $\theta^* < 0$. Setting $P = D^TBD$ and $Q = D^TAD$ and taking into account (*), we see that the matrices P, Q satisfy the premise in (!), whence, by (!), there exists a vector e such that $0 \leq e^T Pe = [De]^T A[De]$ and $0 > e^T Qe = [De]^T B[De]$, which contradicts the premise of the S-Lemma.

It remains to prove (!). Given P and Q as in (!), note that Q , as every symmetric matrix, admits a representation

$$Q = U^T \Lambda U$$

with an orthonormal U and a diagonal Λ . Note that $\theta \equiv \text{Tr}(\Lambda) = \text{Tr}(Q) < 0$. Now let ξ be a random n -dimensional vector with independent entries taking values ± 1 with probabilities $1/2$. We have

$$[U^T \xi]^T Q [U^T \xi] = [U^T \xi]^T U^T \Lambda U [U^T \xi] = \xi^T \Lambda \xi = \text{Tr}(\Lambda) = \theta \quad \forall \xi,$$

while

$$[U^T \xi]^T P [U^T \xi] = \xi^T [UPU^T] \xi,$$

and the expectation of the latter quantity over ξ is clearly $\text{Tr}(UPU^T) = \text{Tr}(P) \geq 0$. Since the expectation is nonnegative, there is at least one realization $\bar{\xi}$ of our random vector ξ such that

$$0 \leq [U^T \bar{\xi}]^T P [U^T \bar{\xi}].$$

We see that the vector $e = U^T \bar{\xi}$ is a required one: $e^T Qe = \theta < 0$ and $e^T Pe \geq 0$. ■

3.5.2 Inhomogeneous S-Lemma

Proposition 3.5.2 [Inhomogeneous S-Lemma] *Consider optimization problem with quadratic objective and a single quadratic constraint:*

$$f_* = \min_x \left\{ f_0(x) \equiv x^T A_0 x + 2b_0^T x + c_0 : f_1(x) \equiv x^T A_1 x + 2b_1^T x + c_1 \leq 0 \right\} \tag{3.5.3}$$

Assume that the problem is strictly feasible and below bounded. Then the Semidefinite relaxation (3.4.5) of the problem is solvable with the optimal value f_ .*

Proof. By Proposition 3.4.1, the optimal value in (3.4.5) can be only $\leq f_*$. Thus, it suffices to verify that (3.4.5) admits a feasible solution with the value of the objective $\geq f_*$, that is, that there exists $\lambda_* \geq 0$ such that

$$\begin{pmatrix} c_0 + \lambda_* c_1 - f_* & b_0^T + \lambda_* b_1^T \\ b_0 + \lambda_* b_1 & A_0 + \lambda_* A_1 \end{pmatrix} \succeq 0. \tag{3.5.4}$$

To this end, let us associate with (3.5.3) a pair of homogeneous quadratic forms of the extended vector of variables $y = (t, x)$, where $t \in \mathbf{R}$, specifically, the forms

$$y^T P y \equiv x^T A_1 x + 2t b_1^T x + c_1 t^2, \quad y^T Q y = -x^T A_0 y - 2t b_0^T x - (c_0 - f_*) t^2.$$

We claim that, first, there exist $\epsilon_0 > 0$ and \bar{y} with $\bar{y}^T P \bar{y} < -\epsilon_0 \bar{y}^T \bar{y}$ and, second, that for every $\epsilon \in (0, \epsilon_0]$ the implication

$$y^T P y \leq -\epsilon y^T y \Rightarrow y^T Q y \leq 0 \quad (3.5.5)$$

holds true. The first claim is evident: by assumption, there exists \bar{x} such that $f_1(\bar{x}) < 0$; setting $\bar{y} = (1, \bar{x})$, we see that $\bar{y}^T P \bar{y} = f_1(\bar{x}) < 0$, whence $\bar{y}^T P \bar{y} < -\epsilon_0 \bar{y}^T \bar{y}$ for appropriately chosen $\epsilon_0 > 0$. To support the second claim, assume that $y = (t, x)$ is such that $y^T P y \leq -\epsilon y^T y$, and let us prove that then $y^T Q y \leq 0$.

- Case 1: $t \neq 0$. Setting $y' = t^{-1}y = (1, x')$, we have $f_1(x') = [y']^T P y' = t^{-2} y^T P y \leq 0$, whence $f_0(x') \geq f_*$, or, which is the same, $[y']^T Q y' \leq 0$, so that $y^T Q y \leq 0$, as required in (3.5.5).
- Case 2: $t = 0$. In this case, $-\epsilon x^T x = -\epsilon y^T y \geq y^T P y = x^T A_1 x$ and $y^T Q y = -x^T A_0 x$, and we should prove that the latter quantity is nonpositive. Assume, on the contrary, that this quantity is positive, that is, $x^T A_0 x < 0$. Then $x \neq 0$ and therefore $x^T A_1 x \leq -\epsilon x^T x < 0$. From $x^T A_1 x < 0$ and $x^T A_0 x < 0$ it follows that $f_1(sx) \rightarrow -\infty$ and $f_0(sx) \rightarrow -\infty$ as $s \rightarrow +\infty$, which contradicts the assumption that (3.5.3) is below bounded. Thus, $y^T Q y \leq 0$.

Our observations combine with \mathcal{S} -Lemma to imply that for every $\epsilon \in (0, \epsilon_0]$ there exists $\lambda = \lambda_\epsilon \geq 0$ such that

$$B \preceq \lambda_\epsilon (A + \epsilon I), \quad (3.5.6)$$

whence, in particular,

$$\bar{y}^T B \bar{y} \leq \lambda_\epsilon \bar{y}^T [A + \epsilon I] \bar{y}.$$

The latter relation, due to $\bar{y}^T A \bar{y} < 0$, implies that λ_ϵ remains bounded as $\epsilon \rightarrow +0$. Thus, we have $\lambda_{\epsilon_i} \rightarrow \lambda_* \geq 0$ as $i \rightarrow \infty$ for a properly chosen sequence $\epsilon_i \rightarrow +0$ of values of ϵ , and (3.5.6) implies that $B \preceq \lambda_* A$. Recalling what are A and B , we arrive at (3.5.4). ■

3.5.3 Approximate \mathcal{S} -Lemma

In general, the \mathcal{S} -Lemma fails to be true when there is more than a single quadratic form in (3.5.2) (that is, when $m > 1$). Similarly, Inhomogeneous \mathcal{S} -Lemma fails to be true for general quadratic quadratically constrained problems with more than a single quadratic constraint. There exists, however, a useful *approximate* version of the Inhomogeneous \mathcal{S} -Lemma in the “multi-constrained” case which is as follows:

Proposition 3.5.3 [Approximate \mathcal{S} -Lemma] *Consider the following quadratic quadratically constrained optimization problem:*

$$\text{Opt} = \max_x \left\{ f(x) = x^T A x + 2b^T x : x^T A_i x \leq c_i, i = 1, \dots, m \right\}, \quad (3.5.7)$$

where $c_i > 0$, $A_i \succeq 0$, $i = 1, \dots, m$ (A can be arbitrary symmetric matrix) and $\sum_{i=1}^m A_i \succ 0$. Let SDP be the optimal value in the Semidefinite relaxation of this problem:

$$\text{SDP} = \min_{\omega, \lambda} \left\{ \omega : \begin{pmatrix} \omega - \sum_{i=1}^m c_i \lambda_i & -b^T \\ -b & \sum_{i=1}^m \lambda_i A_i - A \end{pmatrix} \succeq 0, \lambda \geq 0 \right\} \quad (3.5.8)$$

(note that the problem of interest is a maximization one, whence the difference between the relaxation and (3.4.5)). Then

$$\text{Opt} \leq \text{SDP} \leq \Theta \text{Opt}, \quad (3.5.9)$$

where $\Theta = 1$ in the case of $m = 1$ and

$$\Theta = 2 \ln \left(6 \sum_{i=1}^m \text{Rank}(A_i) \right)$$

in the case of $m > 1$. Moreover, in the latter case there exists x_* such that

$$\begin{aligned} b^T x_* &\geq 0, \\ x_*^T A x_* + 2b^T x_* &\geq \text{SDP}, \\ x_*^T A_i x_* &\leq \Theta c_i, i = 1, \dots, m. \end{aligned} \quad (3.5.10)$$

Proof. The case of $m = 1$ is given by the Inhomogeneous \mathcal{S} -Lemma. Thus, let $m > 1$.

We clearly have

$$\begin{aligned} \text{Opt} &= \max_{t,x} \left\{ x^T A x + 2tb^T x : t^2 \leq 1, x^T A_i x \leq c_i, i = 1, \dots, m \right\} \\ &= \max_{z=(t,x)} \left\{ z^T B z : z^T B_i z \leq c_i, i = 0, 1, \dots, m \right\}, \\ B &= \left[\begin{array}{c|c} b^T & \\ \hline b & A \end{array} \right], \\ B_0 &= \left[\begin{array}{c|c} 1 & \\ \hline & \end{array} \right], \\ B_i &= \left[\begin{array}{c|c} & \\ \hline & A_i \end{array} \right], i = 1, \dots, m, \\ c_0 &= 1. \end{aligned} \quad (3.5.11)$$

Note that (3.5.8) is nothing but the semidefinite dual of the semidefinite program

$$\max_Z \{ \text{Tr}(BZ) : \text{Tr}(B_i Z) \leq c_i, i = 0, 1, \dots, m, Z \succeq 0 \}. \quad (3.5.12)$$

Since $c_i > 0$ for $i \geq 0$, (3.5.12) is strictly feasible, and since $\sum_{i=0}^m B_i \succ 0$, the feasible set of the problem is bounded (so that the problem is solvable). Since (3.5.12) is strictly feasible and bounded, by Semidefinite Duality Theorem we have

$$\text{SDP} = \max_Z \{ \text{Tr}(BZ) : \text{Tr}(B_i Z) \leq c_i, i = 0, 1, \dots, m, Z \succeq 0 \}. \quad (3.5.13)$$

2⁰. Let Z_* be an optimal solution to (3.5.12). Let us set

$$\begin{aligned}\hat{B} &= Z_*^{1/2} B Z_*^{1/2} \\ \hat{B}_i &= Z_*^{1/2} B_i Z_*^{1/2}, \quad i = 0, \dots, m,\end{aligned}$$

and let $\hat{B} = U^T D U$ be the eigenvalue decomposition of \hat{B} (so that U is orthogonal and D is diagonal). Finally, let us set $D_i = U \hat{B}_i U^T$. Thus, we have arrived at the symmetric matrices D, D_0, D_1, \dots, D_m such that

- 1) $D = U Z_*^{1/2} B Z_*^{1/2} U^T$ is diagonal, and $\text{Tr}(D) = \text{Tr}(Z_* B) = \text{SDP}$;
- 2) For $i = 0, 1, \dots, m$, the matrices $D_i = U Z_*^{1/2} B_i Z_*^{1/2} U^T$ are symmetric positive semidefinite, $\text{Rank}(D_i) \leq \text{Rank}(B_i)$, and $\text{Tr}(D_i) = \text{Tr}(Z_* B_i) \leq c_i$.

Now let ξ be a random vector with independent coordinates taking values ± 1 with probability $1/2$, and let $\eta = Z_*^{1/2} U^T \xi$. Observe that

$$\begin{aligned}(a) \quad \eta^T B \eta &\equiv \xi^T D \xi \\ &\equiv \text{SDP} && [\text{by 1}]) \\ (b) \quad \mathbf{E} \left\{ \eta^T B_i \eta \right\} &= \mathbf{E} \left\{ \xi^T D_i \xi \right\} = \text{Tr}(D_i) && (3.5.14) \\ &\leq c_i, \quad i = 0, 1, \dots, m && [\text{by 2}]\end{aligned}$$

3⁰. We claim that from (3.5.14.b) it follows that

$$\begin{aligned}(a) \quad \text{Prob} \left\{ \eta^T B_0 \eta > 1 \right\} &\leq \frac{2}{3} \\ (b) \quad \text{Prob} \left\{ \eta^T B_i \eta > \theta c_i \right\} &\leq \text{Rank}(A_i) \exp\{-\theta/2\}, \quad i = 1, \dots, m && (3.5.15)\end{aligned}$$

Indeed, $\eta^T B_0 \eta = \xi^T D_0 \xi$. Note that $D_0 = d_0 d_0^T$ for certain vector d_0 (since $B_0 = b_0 b_0^T$ for certain b_0). Besides this, $\text{Tr}(D_0) \leq c_0 = 1$ by (3.5.14.b). Thus,

$$\eta^T B_0 \eta = \left(\sum_j p_j \epsilon_j \right)^2,$$

where deterministic p_j satisfy $\sum_j p_j^2 \leq 1$, and ϵ_j are independent random variables taking values ± 1 with probabilities $1/2$. Now (3.5.15.a) is given by the following fact [4]:

With p_j and ϵ_j as above, one has $\text{Prob} \left\{ \left| \sum_j p_j \epsilon_j \right| \leq 1 \right\} \geq \frac{1}{3}$.

To verify (3.5.15.b), let us fix $i \geq 1$. Since D_i is positive semidefinite along with B_i , we have

$$D_i = \sum_{j=1}^k d_j d_j^T \quad [k = \text{Rank}(D_i) \leq \text{Rank}(B_i) = \text{Rank}(A_i)].$$

By Bernstein's Inequality (see the proof of Proposition 2.4.1), we have

$$\text{Prob} \left\{ |d_j^T \xi| \geq \sqrt{\theta} \|d_j\|_2 \right\} \leq 2 \exp\{-\theta/2\}.$$

In the case of $\xi^T D_i \xi \geq \theta \sum_{j=1}^k \|d_j\|_2^2$ we clearly have $\xi^T d_j d_j^T \xi \geq \theta \|d_j\|_2^2$ for certain (depending on ξ) value of j , so that

$$\begin{aligned} \text{Prob} \left\{ \xi^T D_i \xi > \theta \sum_{j=1}^k \|d_j\|_2^2 \right\} &\leq \sum_{j=1}^k \text{Prob} \left\{ |d_j^T \xi| \geq \sqrt{\theta} \|d_j\|_2 \right\} \\ &\leq 2k \exp\{-\theta/2\}. \end{aligned}$$

The resulting inequality implies (3.5.15.b) due to the facts that $\eta^T B_i \eta = \xi^T D_i \xi$ and that

$$\sum_{j=1}^k \|d_j\|_2^2 = \text{Tr}(\sum_j d_j d_j^T) = \text{Tr}(D_i) \leq c_i.$$

4^0 . Let $K = \sum_{i=1}^m \text{Rank}(A_i)$. For every $\theta > \Theta = 2 \ln(6K)$ we have $\theta \geq 1$ and $\frac{2}{3} + 2K \exp\{-\theta/2\} < 1$. In view of the latter fact and (3.5.15), there exists a realization $\bar{\eta} = (\bar{t}, \bar{x})$ of η such that

$$\begin{aligned} (a) \quad \bar{t}^2 &\equiv \bar{\eta}^T B_0 \bar{\eta} \leq 1, \\ (b) \quad \bar{x}^T A_i \bar{x} &\equiv \bar{\eta}^T B_i \bar{\eta} \leq \theta c_i, \quad i = 1, \dots, m. \end{aligned} \tag{3.5.16}$$

while

$$\bar{x}^T A \bar{x} + 2\bar{t}\bar{b}^T \bar{x} = \bar{\eta}^T B \bar{b} \eta = \text{SDP}$$

by (3.5.14.a). Replacing, if necessary, \bar{t} with $-\bar{t}$ and \bar{x} with $-\bar{x}$, we ensure the validity of (3.5.16.b) along with the relation

$$\bar{x}^T A \bar{x} + \underbrace{2\bar{b}^T \bar{x}}_{\geq 0} \geq \text{SDP}. \tag{3.5.17}$$

Since $\sum_i A_i \succ 0$, relations (3.5.16) imply that (\bar{t}, \bar{x}) remain bounded as $\theta \rightarrow +\Theta$, whence (3.5.16) and (3.5.17) are valid for properly chosen \bar{t}, \bar{x} and $\theta = \Theta$; setting $x_* = \bar{x}$, we arrive at (3.5.10). By (3.5.10), $\Theta^{-1/2} x_*$ is a feasible solution of (3.5.7) and

$$\Theta^{-1} x_*^T A x_* + 2\Theta^{-1/2} b^T x_* \geq \Theta^{-1} \text{SDP}, \tag{3.5.18}$$

whence $\text{Opt} \geq \Theta^{-1} \text{SDP}$. ■

3.5.3.1 Application: Approximating the Affinely Adjustable Robust Counterpart of an Uncertain Linear Programming problem

The notion of Affinely Adjustable Robust Counterpart (AARC) of uncertain LP was introduced and motivated in Section 2.4.5. As applied to uncertain LP

$$\mathcal{LP} = \left\{ \min_x \left\{ c^T[\zeta]x : A[\zeta]x - b[\zeta] \geq 0 \right\} : \zeta \in \mathcal{Z} \right\} \tag{3.5.19}$$

affinely parameterized by perturbation vector ζ and with variables x_j allowed to be affine functions of $P_j \zeta$:

$$x_j = \mu_j + \nu_j^T P_j \zeta, \tag{3.5.20}$$

the AARC is the following semi-infinite optimization program in variables t, μ_j, ν_j :

$$\min_{t, \{\mu_j, \nu_j\}_{j=1}^n} \left\{ t : \begin{array}{l} \sum_j c_j[z][\mu_j + \nu_j^T P_j \zeta] \leq t \quad \forall \zeta \in \mathcal{Z} \\ \sum_j [\mu_j + \nu_j^T P_j] A_j[\zeta] - b[\zeta] \geq 0 \quad \forall \zeta \in \mathcal{Z} \end{array} \right\} \quad (\text{AARC})$$

It was explained that in the case of *fixed recourse* ($c_j[\zeta]$ and $A_j[\zeta]$ are independent of ζ for all j for which x_j is adjustable, that is, $P_j \neq 0$), (AARC) is equivalent to an explicit conic quadratic program, provided that the perturbation set \mathcal{Z} is CQR with strictly feasible CQR. In fact CQ-representability plays no crucial role here (see Remark 2.4.1); in particular, when \mathcal{Z} is SDR with a strictly feasible SDR, (AARC), in the case of fixed recourse, is equivalent to an explicit semidefinite program. What indeed plays a crucial role is the assumption of fixed recourse; it can be shown that when this assumption does not hold, (AARC) can be computationally intractable. Our current goal is to demonstrate that even in this difficult case (AARC) admits a “tight” computationally tractable approximation, provided that \mathcal{Z} is an intersection of ellipsoids centered at the origin:

$$\mathcal{Z} = \mathcal{Z}_\rho \equiv \left\{ \zeta : \zeta^T Q_i \zeta \leq \rho^2, i = 1, \dots, m \right\} \quad \left[Q_i \succeq 0, \sum_i Q_i \succ 0 \right] \quad (3.5.21)$$

Indeed, since $A_j[\zeta]$ are affine in ζ , every semi-infinite constraint in (AARC) is of the form

$$\zeta^T A[z] \zeta + 2b^T[z] \zeta \leq c[z] \quad \forall \zeta \in \mathcal{Z}_\rho \quad (3.5.22)$$

where $z = (t, \{\mu_j, \nu_j\}_{j=1}^n)$ is the vector of variables in (AARC), and $A[z]$, $b[z]$, $c[z]$ are affine in z matrix, vector and scalar. Applying Approximate \mathcal{S} -Lemma, a sufficient condition for (3.5.22) to be valid for a given z is the relation

$$\min_{\omega, \lambda} \left\{ \omega : \begin{pmatrix} \omega - \sum_{i=1}^m \rho^2 \lambda_i & -b^T[z] \\ -b[z] & \sum_{i=1}^m \lambda_i Q_i - A[z] \end{pmatrix} \succeq 0, \lambda \geq 0 \right\} \leq c[z],$$

or, which is the same, the possibility to extend z , by properly chosen λ , to a solution to the system of constraints

$$\lambda \geq 0, \begin{pmatrix} c[z] - \sum_{i=1}^m \rho^2 \lambda_i & -b^T[z] \\ -b[z] & \sum_{i=1}^m \lambda_i Q_i - A[z] \end{pmatrix}, \quad (3.5.23)$$

which is a system of LMIs in variables (z, λ) . Replacing every one of the semi-infinite constraints in (AARC) with corresponding system (3.5.23), we end up with an explicit semidefinite program which is a “conservative approximation” of (AARC): both problems have the same objective, and the z -component of a feasible solution to the approximation is feasible solution of (AARC). At the same time, the approximation is tight up to the quite moderate factor

$$\theta = \sqrt{2 \ln \left(6 \sum_{i=1}^m \text{Rank}(Q_i) \right)} :$$

whenever z cannot be extended to a feasible solution of the approximation, the “moreover” part of the Approximate \mathcal{S} -Lemma says that z becomes infeasible for (AARC) after the original level of perturbations is increased by the factor θ .

3.6 Semidefinite Relaxation and Chance Constraints

3.6.1 Situation and goal

3.6.1.1 Chance constraints

We already have touched on two different occasions (Sections 2.4, 3.4.3) the important area of *optimization under uncertainty* — solving optimization problems with partially unknown (“uncertain”) data. However, for the time being, we have dealt solely with *uncertain-but-bounded* data perturbations, that is, data perturbations running through a given *uncertainty set*, and with a specific interpretation of an uncertainty-affected constraint: we wanted from the candidate solutions (non-adjustable or adjustable alike) to satisfy the constraints for *all* realizations of the data perturbations from the uncertainty set; this was called *Robust Optimization paradigm*. In Optimization there exists another approach to treating data uncertainty, the *Stochastic Optimization* approach historically by far preceding the RO one. With the Stochastic Optimization approach, the data perturbations are assumed to be *random* with completely or partially known probability distribution. In this situation, a natural way to handle data uncertainty is to pass to *chance constrained* forms of the uncertainty-affected constraints¹⁸, namely, to replace a constraint

$$f(x, \zeta) \leq 0$$

(x is the desicion vector, ζ stands for data perturbation) with the constraint

$$\text{Prob}_{\zeta \sim P} \{\zeta : f(x, \zeta) \leq 0\} \leq 1 - \epsilon,$$

where $\epsilon \ll 1$ is a given tolerance, and P is the distribution of ζ . This formulation tacitly assumes that the distribution of ζ is known exactly, which typically is *not* the case in reality. In real life, even if we have reasons to believe that the data perturbations are random (which by itself is a “big if”), we usually are not informed enough to point out their distribution exactly. Instead, we usually are able to point out a *family* \mathcal{P} of probability distributions which contains the true distribution of ζ . In this case, we usually pass to *ambiguously chance constrained* version of the uncertain constraint:

$$\sup_{P \in \mathcal{P}} \text{Prob}_{\zeta \sim P} \{\zeta : f(x, \zeta) \leq 0\} \geq 1 - \epsilon. \quad (3.6.1)$$

Chance constraints is a quite old notion (going back to mid-1950’s); over the years, they were subject of intensive and fruitful research of numerous excellent scholars worldwide. With all due respect to this effort, chance constraints, even as simply-looking as scalar linear inequalities with randomly perturbed coefficients, remain quite challenging computationally¹⁹. The reason is twofold:

- Checking the validity of a chance constraint at a given point x , even with exactly known distribution of ζ , requires multi-dimensional (provided ζ is so) integration, which typically is a computationally intractable task. Of course, one can replace precise integration

¹⁸By adding slack variable t and replacing minimization of the true objective $f(x)$ with minimizing t under added constraint $f(x) \leq t$, we always can assume that the objective is linear and certain — not affected by data uncertainty. By this reason, we can restrict ourselves with the case where the only uncertainty-affected components of an optimization problem are the constraints.

¹⁹By the way, this is perhaps the most important argument in favour of RO: as we have seen in Section 2.4, the RO approach, at least in the case of non-adjustable uncertain Linear Programming, results in computationally tractable (provided the uncertainty set is so) Robust Counterpart of the uncertain problem of interest.

by Monte Carlo simulation — by generating a long sample of independent realizations ζ^1, \dots, ζ^N of ζ and replacing the true probability with its empirical approximation — the frequency of the event $f(x, \zeta^t) \leq 0$ in the sample. However, this approach is unapplicable to the case of an ambiguous chance constraint and, in addition, does not work when ϵ is “really small,” like 10^{-6} or less. The reason is that in order to decide reliably from simulations that the probability of the event $f(x, \zeta) > 0$ indeed is $\leq \epsilon$, the sample size N should be of order of $1/\epsilon$ and thus becomes impractically large when ϵ is “really small.”

- Another potential difficulty with chance constraint is that its feasible set can be non-convex already for a pretty simple — just affine in x — function $f(x, \zeta)$, which makes the optimization of objective over such a set highly problematic.

Essentially, the only generic situation where neither one of the above two difficulties occur is the case of scalar linear constraint

$$\sum_{i=1}^n \zeta_i x_i \leq 0$$

with Gaussian $\zeta = [\zeta_1; \dots; \zeta_N]$. Assuming that the distribution of ζ is known exactly and denoting by μ , Σ the expectation and the covariance matrix of ζ , we have

$$\text{Prob}\{\sum_i \zeta_i x_i \leq 0\} \geq 1 - \epsilon \Leftrightarrow \sum_i \mu_i \zeta_i + \Phi^{-1}(\epsilon) \sqrt{x^T \Sigma x} \leq 0,$$

where $\Phi^{-1}(\cdot)$ is the *inverse error function* given by:

$$\int_{\Phi^{-1}(\epsilon)}^{\infty} \frac{1}{\sqrt{2\pi}} \exp\{-s^2/2\} ds = \epsilon.$$

When $\epsilon \leq 1/2$, the above “deterministic equivalent” of the chance constraint is an explicit Conic Quadratic inequality.

3.6.1.2 Safe tractable approximations of chance constraints

When a chance constraint “as it is” is computationally intractable, one can look for the “second best thing” — a system \mathcal{S} of efficiently computable convex constraints in variable x and perhaps additional slack variables u which “safely approximates” the chance constraint, meaning that whenever x can be extended, by a properly chosen u , to a feasible solution of \mathcal{S} , x is feasible for the chance constraint of interest. For a survey of state-of-the-art techniques for safe tractable approximations of chance constraints, we refer an interested reader to [22, 5] and references therein. In the sequel, we concentrate on a particular technique of this type, originating from [7, 6] and heavily exploiting semidefinite relaxation; the results to follow are taken from [22].

3.6.1.3 Situation and goal

In the sequel, we focus on the chance constrained form of a *quadratically perturbed scalar linear constraint*. A generic form of such a chance constraint is

$$\text{Prob}_{\zeta \sim P} \{\text{Tr}(WZ[\zeta]) \leq 0\} \geq 1 - \epsilon \quad \forall P \in \mathcal{P}, \quad Z[\zeta] = \begin{bmatrix} 1 & | & \zeta^T \\ \zeta & | & \zeta \zeta^T \end{bmatrix} \quad \forall P \in \mathcal{P}, \quad (3.6.2)$$

where the data perturbations $\zeta \in \mathbf{R}^d$, and the symmetric matrix W is affine in the decision variables; we lose nothing by assuming that W itself is the decision variable.

We start with the description of \mathcal{P} . Specifically, we assume that our a priori information on the distribution P of the uncertain data can be summarized as follows:

- $\mathcal{P}.1$ We know that the marginals P_i of P (i.e., the distributions of the entries ζ_i in ζ) belong to given families \mathcal{P}_i of probability distributions on the axis;
- $\mathcal{P}.2$ The matrix $V_P = \mathbf{E}_{\zeta \sim P}\{Z[\zeta]\}$ of the first and the second moments of P is known to belong to a given convex closed subset \mathcal{V} of the positive semidefinite cone;
- $\mathcal{P}.3$ P is supported on a set S given by a finite system of quadratic constraints:

$$S = \{\zeta : \text{Tr}(A_\ell Z[\zeta]) \leq 0, 1 \leq \ell \leq L\}$$

The above assumptions model well enough a priori information on uncertain data in typical applications of decision-making origin.

3.6.2 Approximating chance constraints via Lagrangian relaxation

The approach, which we in the sequel refer to as *Lagrangian approximation*, implements the idea as follows. Assume that given W , we have a systematic way to generate pairs $(\gamma(\cdot) : \mathbf{R}^d \rightarrow \mathbf{R}, \theta > 0)$ such that (I) $\gamma(\cdot) \geq 0$ in S , (II) $\gamma(\cdot) \geq \theta$ in the part of S where $\text{Tr}(WZ[\zeta]) \geq 0$, and (c) we have at our disposal a functional $\Gamma[\gamma]$ such that $\Gamma[\gamma] \geq \Gamma_*[\gamma] := \sup_{P \in \mathcal{P}} \mathbf{E}_{\zeta \sim P}\{\gamma(\zeta)\}$. Since $\gamma(\cdot)$ is nonnegative at the support of ζ and is $geq \theta$ in the part of this support where the body $\text{Tr}(WZ[\zeta])$ of the chance constraint is positive, we clearly have $\Gamma[\gamma] \geq \mathbf{E}_{\zeta \sim P}\{\gamma(\zeta)\} \geq \theta \text{Prob}_{\zeta \sim P}\{\text{Tr}(WZ[\zeta]) < 0\}$, so that the condition

$$\Gamma[\gamma] \leq \theta\epsilon$$

clearly is sufficient for the validity of (3.6.2), and we can further optimize this sufficient condition over the pairs γ, θ produced by our hypothetic mechanism.

The question is, how to build the required mechanism, and here is an answer. Let us start with building $\Gamma[\cdot]$. Under our assumptions on \mathcal{P} , the most natural family of functions $\gamma(\cdot)$ for which one can bound from above $\Gamma_*[\gamma]$ is comprised of functions of the form

$$\gamma(\zeta) = \text{Tr}(QZ[\zeta]) + \sum_{i=1}^d \gamma_i(\zeta_i) \quad (3.6.3)$$

with $Q \in \mathbf{S}^{d+1}$. For such a γ , one can set

$$\Gamma[\gamma] = \sup_{V \in \mathcal{V}} \text{Tr}(QV) + \sum_{i=1}^d \sup_{P_i \in \mathcal{P}_i} \int \gamma_i(\zeta_i) dP_i(\zeta_i)$$

Further, the simplest way to ensure (I) is to use *Lagrangian relaxation*, specifically, to require from the function $\gamma(\cdot)$ given by (3.6.3) to be such that with properly chosen $\mu_\ell \geq 0$ one has

$$\text{Tr}(QZ[\zeta]) + \sum_{i=1}^d \gamma_i(\zeta_i) + \sum_{\ell=1}^L \mu_\ell \text{Tr}(A_\ell Z[\zeta]) \geq 0 \quad \forall \zeta \in \mathbf{R}^d.$$

In turn, the simplest way to ensure the latter relation is to impose on $\gamma_i(\zeta_i)$ the restrictions

- (a) $\gamma_i(\zeta_i) \geq p_i \zeta_i^2 + 2q_i \zeta_i + r_i \quad \forall \zeta_i \in \mathbf{R}$,
- (b) $\text{Tr}(QZ[\zeta]) + \sum_{i=1}^d [p_i \zeta_i^2 + 2q_i \zeta_i + r_i] + \sum_{\ell=1}^L \mu_\ell \text{Tr}(A_\ell Z[\zeta]) \geq 0 \quad \forall \zeta \in \mathbf{R}^d$;

note that (b) reduces to the LMI

$$Q + \left[\begin{array}{c|c} \sum_i r_i & q^T \\ \hline q & \text{Diag}\{p\} \end{array} \right] + \sum_{\ell=1}^L \mu_\ell A_\ell \succeq 0 \quad (3.6.5)$$

in variables $p = [p_1; \dots; p_d]$, $q = [q_1; \dots; q_d]$, $r = [r_1; \dots; r_d]$, Q , $\{\mu_\ell \geq 0\}_{\ell=1}^L$.

Similarly, a sufficient condition for (II) is the existence of $p', q', r' \in \mathbf{R}^d$ and nonnegative ν_ℓ such that

$$\begin{aligned} (a) \quad & \gamma_i(\zeta_i) \geq p'_i \zeta_i^2 + 2q'_i \zeta_i + r'_i \forall \zeta_i \in \mathbf{R}, \\ (b) \quad & \text{Tr}(QZ[\zeta]) + \sum_{i=1}^d [p'_i \zeta_i^2 + 2q'_i \zeta_i + r'_i] + \sum_{\ell=1}^L \nu_\ell \text{Tr}(A_\ell Z[\zeta]) - \text{Tr}(WZ[\zeta]) \geq \theta \forall \zeta \in \mathbf{R}^d, \end{aligned} \quad (3.6.6)$$

with (b) reducing to the LMI

$$Q + \left[\begin{array}{c|c} \sum_i r'_i - \theta & [q']^T \\ \hline q' & \text{Diag}\{p'\} \end{array} \right] + \sum_{\ell=1}^L \nu_\ell A_\ell - W \succeq 0 \quad (3.6.7)$$

in variables W , $p', q', r' \in \mathbf{R}^d$, Q , $\{\nu_i \geq 0\}_{i=1}^L$ and $\theta > 0$. Finally, observe that under the restrictions (3.6.4.a), (3.6.6.a), the best – resulting in the smallest possible $\Gamma[\gamma]$ – choice of $\gamma_i(\cdot)$ is

$$\gamma_i(\zeta_i) = \max[p_i \zeta_i^2 + 2q_i \zeta_i + r_i, p'_i \zeta_i^2 + 2q'_i \zeta_i + r'_i].$$

We have arrived at the following result:

Proposition 3.6.1 *Let (\mathcal{S}) be the system of constraints in variables W , $p, q, r, p', q', r' \in \mathbf{R}^d$, $Q \in \mathbf{S}^{d+1}$, $\{\nu_\ell \in \mathbf{R}\}_{\ell=1}^L$, $\theta \in \mathbf{R}$, $\{\mu_\ell \in \mathbf{R}\}_{\ell=1}^L$ comprised of the LMIs (3.6.5), (3.6.7) augmented by the constraints*

$$\mu_\ell \geq 0 \forall \ell, \nu_\ell \geq 0 \forall \ell, \theta > 0 \quad (3.6.8)$$

and

$$\sup_{V \in \mathcal{V}} \text{Tr}(QV) + \sum_{i=1}^d \sup_{P_i \in \mathcal{P}_i} \int \max[p_i \zeta_i^2 + 2q_i \zeta_i + r_i, p'_i \zeta_i^2 + 2q'_i \zeta_i + r'_i] dP_i \leq \epsilon \theta. \quad (3.6.9)$$

(\mathcal{S}) is a system of convex constraints which is a safe approximation of the chance constraint (3.6.2), meaning that whenever W can be extended to a feasible solution of (\mathcal{S}) , W is feasible for the ambiguous chance constraint aneq500), \mathcal{P} being given by P.1-3. This approximation is tractable, provided that the suprema in (3.6.9) are efficiently computable.

Note that the strict inequality $\theta > 0$ in (3.6.8) can be expressed by the LMI $\begin{bmatrix} \theta & 1 \\ 1 & \lambda \end{bmatrix} \succeq 0$.

3.6.2.1 Illustration I

Consider the situation as follows: there are $d = 15$ assets with yearly returns $r_i = 1 + \mu_i + \sigma_i \zeta_i$, where μ_i is the expected profit of i -th return, σ_i is the return's variability, and ζ_i is random factor with zero mean supported on $[-1, 1]$. The quantities μ_i , σ_i used in our illustration are shown on the left plot on figure 3.1. The goal is to distribute \$1 between the assets in order to maximize the value-at-1% risk (the lower 1%-quantile) of the yearly profit. This is the ambiguously chance constrained problem

$$\text{Opt} = \max_{t,x} \left\{ t : \text{Prob}_{\zeta \sim P} \left\{ \sum_{i=1}^{15} \mu_i x_i + \sum_{i=1}^{15} \zeta_i \sigma_i x_i \geq t \right\} \geq 0.99 \forall P \in \mathcal{P}, x \geq 0, \sum_{i=1}^{15} x_i = 1 \right\} \quad (3.6.10)$$

Hypothesis	Approximation	Guaranteed profit-at-1%-risk
A	Bernstein	0.0552
B, C	Lagrangian	0.0101

Table 3.1: Optimal values in various approximations of (3.6.10).

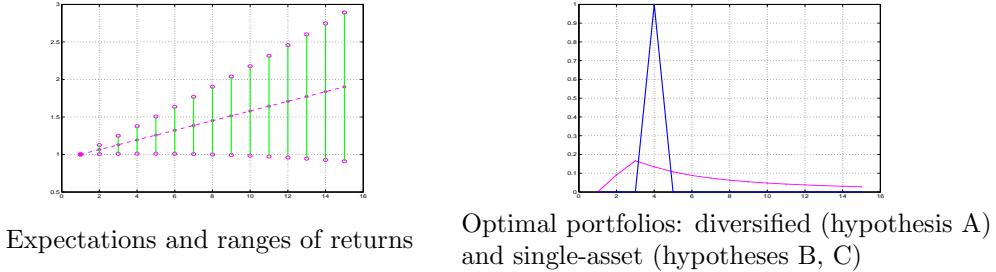


Figure 3.1: Data and results for portfolio allocation.

Consider three hypotheses A, B, C about \mathcal{P} . In all of them, ζ_i are zero mean and supported on $[-1, 1]$, so that the domain information is given by the quadratic inequalities $\zeta_i^2 \leq 1$, $1 \leq i \leq 15$; this is exactly what is stated by C. In addition, A says that ζ_i are independent, and B says that the covariance matrix of ζ is proportional to the unit matrix. Thus, the sets \mathcal{V} associated with the hypotheses are, respectively, $\{V \in \mathbf{S}_+^{d+1} : V_{ii} \leq V_{00} = 1, V_{ij} = 0, i \neq j\}$, $\{V \in \mathbf{S}_+^{d+1} : 1 = V_{00} \geq V_{11} = V_{22} = \dots = V_{dd}, V_{ij} = 0, i \neq j\}$, and $\{V \in \mathbf{S}_+^{d+1} : V_{ii} \leq V_{00} = 1, V_{0j} = 0, 1 \leq j \leq d\}$, where \mathbf{S}_+^k is the cone of positive semidefinite symmetric $k \times k$ matrices. Solving the associated safe tractable approximations of the problem, specifically, the Bernstein approximation in the case of A, and the Lagrangian approximations in the cases of B, C, we arrive at the results displayed in table 3.1 and on figure 3.1.

Note that in our illustration, the (identical to each other) single-asset portfolios yielded by the Lagrangian approximation under hypotheses B, C are *exactly optimal* under circumstances. Indeed, on a closest inspection, there exists a distribution P_* compatible with hypothesis B (and therefore – with C as well) such that the probability of “crisis,” where all ζ_i simultaneously are equal to -1 , is ≥ 0.01 . It follows that under hypotheses B, C, the worst-case, over $P \in \mathcal{P}$, profit at 1% risk of *any* portfolio cannot be better than the profit of this portfolio in the case of crisis, and the latter quantity is maximized by the single-asset portfolio depicted on figure 3.1. Note that the Lagrangian approximation turns out to be “intelligent enough” to discover this phenomenon and to infer its consequences. A couple of other instructive observations is as follows:

- the diversified portfolio yielded by the Bernstein approximation²⁰ in the case of crisis exhibits negative profit, meaning that under hypotheses B, C its worst-case profit at 1% risk is negative;
- assume that the yearly returns are observed on a year-by-year basis, and the year-by-year

²⁰See, e.g., [22]. This is a specific safe tractable approximation of affinely perturbed linear scalar chance constraint which heavily exploits the assumption that the entries ζ_i in ζ are mutually independent.

realizations of ζ are independent and identically distributed. It turns out that it takes over 100 years to distinguish, with reliability 0.99, between hypothesis A and the “bad” distribution P_* via the historical data.

To put these observations into proper perspective, note that it is extremely time-consuming to identify, to reasonable accuracy and with reasonable reliability, a multi-dimensional distribution directly from historical data, so that in applications one usually postulates certain parametric form of the distribution with a relatively small number of parameters to be estimated from the historical data. When $\dim \zeta$ is large, the requirement on the distribution to admit a low-dimensional parameterization usually results in postulating some kind of independence. While in some applications (e.g., in telecommunications) this independence in many cases can be justified via the “physics” of the uncertain data, in Finance and other decision-making applications postulating independence typically is an “act of faith” which is difficult to justify experimentally, and we believe a decision-maker should be well aware of the dangers related to these “acts of faith.”

3.6.3 A Modification

When building safe tractable approximation of an affinely perturbed scalar linear chance constraint

$$\forall p \in \mathcal{P} : \text{Prob}_{\zeta \sim P} \left\{ w_0 + \sum_{i=1}^d \zeta_i w_i \leq 0 \right\} \geq 1 - \epsilon, \quad (3.6.11)$$

one, essentially, is interested in efficient bounding from above the quantity

$$p(w) = \sup_{P \in \mathcal{P}} \mathbf{E}_{\zeta \sim P} \left\{ f(w_0 + \sum_{i=1}^d \zeta_i w_i) \right\}$$

for a very specific function $f(s)$, namely, equal to 0 when $s \leq 0$ and equal to 1 when $s > 0$. There are situations when we are interested in bounding similar quantity for other functions f , specifically, piecewise linear convex function $f(s) = \max_{1 \leq j \leq J} [a_i + b_i s]$, see, e.g., [10]. Here again one can use Lagrange relaxation, which in fact is able to cope with a more general problem of bounding from above the quantity

$$\Psi[W] = \sup_{P \in \mathcal{P}} \mathbf{E}_{\zeta \sim P} \{f(W, \zeta)\}, \quad f(W, \zeta) = \max_{1 \leq j \leq J} \text{Tr}(W^j Z[\zeta]);$$

here the matrices $W^j \in \mathbf{S}^{d+1}$ are affine in the decision variables and $W = [W^1, \dots, W^J]$. Specifically, with the assumptions P.1-3 in force, observe that if, for a given W , a matrix $Q \in \mathbf{S}^{d+1}$ and vectors $p^j, q^j, r^j \in \mathbf{R}^d$, $1 \leq j \leq J$ are such that the relations

$$\text{Tr}(QZ[\zeta]) + \sum_{i=1}^d [p_i^j \zeta_i^2 + 2q_i^j \zeta_i + r_i^j] \geq \text{Tr}(W^j Z[\zeta]) \quad \forall \zeta \in S \quad (I_j)$$

take place for $1 \leq j \leq J$, then the function

$$\gamma(\zeta) = \text{Tr}(QZ[\zeta]) + \sum_{i=1}^d \max_{1 \leq j \leq J} [p_i^j \zeta_i^2 + 2q_i^j \zeta_i + r_i^j] \quad (3.6.12)$$

upper-bounds $f(W, \zeta)$ when $\zeta \in S$, and therefore the quantity

$$F(W, Q, p, q, r) = \sup_{V \in \mathcal{V}} \text{Tr}(QV) + \sum_{i=1}^d \sup_{P_i \in \mathcal{P}_i} \int \max_{1 \leq j \leq J} [p_i^j \zeta_i^2 + 2q_i^j \zeta_i + r_i^j] dP_i(\zeta_i) \quad (3.6.13)$$

is an upper bound on $\Psi[W]$. Using Lagrange relaxation, a sufficient condition for the validity of (I_j) , $1 \leq j \leq J$, is the existence of nonnegative $\mu_{j\ell}$ such that

$$Q + \left[\begin{array}{c|c} \sum_i r_i^j & [q^j]^T \\ \hline q^j & \text{Diag}\{p^j\} \end{array} \right] - W^j + \sum_{\ell=1}^L \mu_{j\ell} A_\ell \succeq 0, \quad 1 \leq j \leq J. \quad (3.6.14)$$

We have arrived at the result as follows:

Proposition 3.6.2 *Consider the system \mathcal{S} of constraints in the variables $W = [W^1, \dots, W^d]$, t and slack variables $Q \in \mathbf{S}^{d+1}$, $\{p^j, q^j, r^j \in \mathbf{R}^d : 1 \leq j \leq J\}$, $\{\mu_{ij} : 1 \leq i \leq d, 1 \leq j \leq J\}$ comprised of LMIs (3.6.14) augmented by the constraints*

$$\begin{aligned} (a) \quad & \mu_{ij} \geq 0, \forall i, j \\ (b) \quad & \max_{V \in \mathcal{V}} \text{Tr}(QV) + \sum_{i=1}^d \sup_{P_i \in \mathcal{P}_i} \int \max_{1 \leq j \leq J} [p_i^j \zeta_i^2 + 2q_i^j \zeta_i + r_i^j] dP_i(\zeta_i) \leq t. \end{aligned} \quad (3.6.15)$$

The constraints in the system are convex; they are efficiently computable, provided that the suprema in (3.6.15.b) are efficiently computable, and whenever W, t can be extended to a feasible solution to \mathcal{S} , one has $\Psi[W] \leq t$. In particular, when the suprema in (3.6.15.b) are efficiently computable, the efficiently computable quantity

$$\text{Opt}[W] = \min_{t, Q, p, q, r, \mu} \{t : W, t, Q, p, q, r, \mu \text{ satisfy } \mathcal{S}\} \quad (3.6.16)$$

is a convex in W upper bound on $\Psi[W]$.

3.6.3.1 Illustration II

Consider a special case of the above situation where all we know about ζ are the marginal distributions P_i of ζ_i with well defined first order moments; in this case, $\mathcal{P}_i = \{P_i\}$ are singletons, and we lose nothing when setting $\mathcal{V} = \mathbf{S}_+^{d+1}$, $S = \mathbf{R}^d$. Let a piecewise linear convex function on the axis:

$$f(s) = \max_{1 \leq j \leq J} [a_j + b_j s]$$

be given, and let our goal be to bound from above the quantity

$$\psi(w) = \sup_{P \in \mathcal{P}} \mathbf{E}_{\zeta \sim P} \{f(\zeta^w)\}, \quad \zeta^w = w_0 + \sum_{i=1}^d \zeta_i w_i.$$

This is a special case of the problem we have considered, corresponding to

$$W^j = \left[\begin{array}{c|ccc} a_j + b_j w_0 & \frac{1}{2} b_j w_1 & \dots & \frac{1}{2} b_j w_d \\ \hline \frac{1}{2} b_j w_1 & & & \\ \vdots & & & \\ \frac{1}{2} b_j w_d & & & \end{array} \right].$$

In this case, system \mathcal{S} from Proposition 3.6.2, where we set $p_i^j = 0$, $Q = 0$, reads

$$\begin{aligned} (a) \quad & 2q_i^j = b_j w_i, \quad 1 \leq i \leq d, \quad q \leq j \leq J, \quad \sum_{i=1}^d r_i^j \geq a_j + b_j w_0, \\ (b) \quad & \sum_{i=1}^d \int \max_{1 \leq j \leq J} [2q_i^j \zeta_i + r_i^j] dP_i(\zeta_i) \leq t \end{aligned}$$

so that the upper bound (3.6.16) on $\sup_{P \in \mathcal{P}} \mathbf{E}_{\zeta \sim P} \{f(\zeta^w)\}$ implies that

$$\text{Opt}[w] = \min_{\{r_i^j\}} \left\{ \sum_{i=1}^d \int \max_{1 \leq j \leq J} [b_j w_i \zeta_i + r_i^j] dP_i(\zeta_i) : \sum_i r_i^j = a_j + b_j w_0, 1 \leq j \leq J \right\} \quad (3.6.17)$$

is an upper bound on $\psi(w)$. A surprising fact is that *in the situation in question* (i.e., when \mathcal{P} is comprised of all probability distributions with given marginals P_1, \dots, P_d), the upper bound $\text{Opt}[w]$ on $\psi(w)$ is equal to $\psi(w)$ [5, Proposition 4.5.4]. This result offers an alternative (and simpler) proof for the following remarkable fact established by Dhaene et al [10]: if f is a convex function on the axis and η_1, \dots, η_d are scalar random variables with distributions P_1, \dots, P_d possessing first order moments, then $\sup_{P \in \mathcal{P}} \mathbf{E}_{\eta \sim P} \{f(\eta_1 + \dots + \eta_d)\}$, \mathcal{P} being the family of all distributions on \mathbf{R}^d with marginals P_1, \dots, P_d , is achieved when η_1, \dots, η_d are comonotone, that is, are deterministic monotone transformation of a single random variable uniformly distributed on $[0, 1]$.

Exercise 3.1 Here you will learn how to verify claims like “distinguishing, with reliability 0.99, between distributions A and B takes at least so much observations.”

1. Problem’s setting. Let P and Q be two probability distributions on the same space Ω with densities $p(\cdot), q(\cdot)$ with respect to some measure μ .

Those with limited experience in measure theory will lose nothing by assuming whenever possible that Ω is the finite set $\{1, \dots, N\}$ and μ is the “counting measure” (the μ -mass of every point from Ω is 1). In this case, the density of a probability distribution P on Ω is just the function $p(\cdot)$ on the N -point set Ω (i.e., N -dimensional vector) with $p(i) = \text{Prob}_{\omega \sim P}\{\omega = i\}$ (that is, $p(i)$ is the probability mass which is assigned by P to a point $i \in \Omega$). In this case (in the sequel, we refer to it as to the *discrete one*), the density of a probability distribution on Ω is just a probabilistic vector from \mathbf{R}^N — a nonnegative vector with entries summing up to 1.

Given an observation $\omega \in \Omega$ drawn at random from one (we do not know in advance from which one) of the distributions P, Q , we want to decide what is the underlying distribution; this is called distinguishing between two simple hypotheses²¹. A (deterministic) decision rule clearly should be as follows: we specify a subset $\Omega_P \subset \Omega$ and accept the hypothesis H_P that the distribution from which ω is drawn is P if and only if $\omega \in \Omega_P$; otherwise we accept the alternative hypothesis H_Q saying that the “actual” distribution is Q .

A decision rule for distinguishing between the hypotheses can be characterized by two error probabilities: ϵ_P (to accept H_Q when H_P is true) and ϵ_Q (to accept H_P when H_Q is true). We clearly have

$$\epsilon_P = \int_{\omega \notin \Omega_P} p(\omega) d\mu(\omega), \quad \epsilon_Q = \int_{\omega \in \Omega_P} q(\omega) d\mu(\omega)$$

Task 1: Prove²² that for every decision rule it holds

$$\epsilon_P + \epsilon_Q \geq \int_{\Omega} \min[p(\omega), q(\omega)] d\mu(\omega).$$

²¹“simple” expresses the fact that every one of the two hypotheses specifies exactly the distribution of ω , in contrast to the general case where a hypothesis can specify only a family of distributions from which the “actual one,” underlying the generation of the observed ω , is chosen.

²²You are allowed to provide the proofs of this and the subsequent claims in the discrete case only.

Prove that this lower bound is achieved for the *maximum likelihood* test, where $\Omega_P = \{\omega : p(\omega) \geq q(\omega)\}$.

Follow-up: Essentially, the only multidimensional case where it is easy to compute the *total error* $\epsilon_P + \epsilon_Q$ of the maximum likelihood test is when P and Q are Gaussian distributions on $\Omega = \mathbf{R}^k$ with common covariance matrix (which we for simplicity set to the unit matrix) and different expectations, say, a and b , so that the densities (taken w.r.t. the usual Lebesque measure $d\mu(\omega_1, \dots, \omega_k) = d\omega_1 \dots d\omega_k$) are $p(\omega) = \frac{1}{(2\pi)^{k/2}} \exp\{-(\omega - a)^T(\omega - a)/2\}$ and $q(\omega) = \frac{1}{(2\pi)^{k/2}} \exp\{-(\omega - b)^T(\omega - b)/2\}$. Prove that in this case the likelihood test reduces to accepting P when $\|\omega - a\|_2 \leq \|\omega - b\|_2$ and accepting Q otherwise.

Prove that the total error of the above test is $2\Phi(\|a - b\|_2/2)$, where

$$\Phi(s) = \frac{1}{\sqrt{2\pi}} \int_s^\infty \exp\{-t^2/2\} dt$$

is the error function.

2. Kullback-Leibler divergence. The question we address next is how to compute the outlined lower bound on $\epsilon_P + \epsilon_Q$ (its explicit representation as an integral does not help much – how could we compute this integral in a multi-dimensional case?)

One of the useful approaches to the task at hand is based on *Kullback-Leibler divergence* defined as

$$\mathcal{H}(p, q) = \int p(\omega) \ln \left(\frac{p(\omega)}{q(\omega)} \right) d\mu(\omega) \quad \left[= \sum_{\omega \in \Omega} p(\omega) \ln \left(\frac{p(\omega)}{q(\omega)} \right) \text{ in the discrete case} \right]$$

In this definition, we set $0 \ln(0/a) = 0$ whenever $a \geq 0$ and $a \ln(a/0) = +\infty$ whenever $a > 0$.

Task 2:

- (a) Compute the Kullback-Leibler divergence between two Gaussian distributions on \mathbf{R}^k with unit covariance matrices and expectations a, b .
- (b) Prove that $\mathcal{H}(p, q)$ is convex function of $(p(\cdot), q(\cdot)) \geq 0$.

Hint: It suffices to prove that the function $t \ln(t/s)$ is a convex function of $t, s \geq 0$. To this end, it suffices to note that this function is the projective transformation $sf(t/s)$ of the (clearly convex) function $f(t) = t \ln(t)$ with the domain $t \geq 0$.

- (c) Prove that when p, q are probability densities, $\mathcal{H}(p, q) \geq 0$.
- (d) Let k be a positive integer and let for $1 \leq s \leq k$ p^s be a probability density on Ω^s taken w.r.t. measure μ^s . Let us define the “direct product” $p^1 \times \dots \times p^k$ of the densities p^s the density of k -element sample $\omega_1, \dots, \omega_k$ with independent ω_s drawn according to p^s , $s = 1, \dots, k$; the product density is taken w.r.t. the measure $\mu = \mu^1 \times \dots \times \mu^k$ (i.e., $d\mu(\omega_1, \dots, \omega_k) = d\mu^1(\omega_1) \dots d\mu^k(\omega_k)$) on the space $\Omega^1 \times \dots \times \Omega^k$ where the sample lives. For example, in the discrete case p^s are probability vectors of dimension N_s , $s = 1, \dots, k$, and $p^1 \times \dots \times p^k$ is the probability vector of the dimension N_1, \dots, N_k with the entries

$$p_{i_1, \dots, i_k} = p_{i_1}^1 p_{i_2}^2 \dots p_{i_k}^k, \quad 1 \leq i_s \leq N_s, \quad 1 \leq s \leq k.$$

Prove that if p^s, q^s are probability densities on Ω_s taken w.r.t. measures μ^s on these spaces, then

$$\mathcal{H}(p^1 \times \dots \times p^k, q^1 \times \dots \times q^k) = \sum_{s=1}^k \mathcal{H}(p^s, q^s).$$

In particular, when all p^s coincide with some p (in this case, we denote $p^1 \times \dots \times p^k$ by $p^{\otimes k}$) and q^1, \dots, q^k coincide with some q , then

$$\mathcal{H}(p^{\otimes k}, q^{\otimes k}) = k\mathcal{H}(p, q).$$

Follow-up: What is Kullback-Leibler divergence between two k -dimensional Gaussian distributions $\mathcal{N}(a, I_k)$ and $\mathcal{N}(b, I_k)$ ($\Omega = \mathbf{R}^k$, $d\mu(\omega_1, \dots, \omega_k) = d\omega_1 \dots d\omega_k$, the density of $\mathcal{N}(a, I_k)$ is $\frac{1}{(2\pi)^{k/2}} \exp\{-(w-a)^T(w-a)/2\}$)?

- (e) Let p, q be probability densities on (Ω, μ) , and let $A \subset \Omega$, $B = \Omega \setminus A$. Let $p_A, p_B = 1 - p_A$ be the p -probability masses of A and B , and let $q_A, q_B = 1 - q_A$ be the q -probability masses of A, B . Prove that

$$\mathcal{H}(p, q) \geq p_A \ln \left(\frac{p_A}{q_A} \right) + p_B \ln \left(\frac{p_B}{q_B} \right).$$

- (f) In the notation from Problem's setting, prove that if the hypotheses H_P, H_Q can be distinguished from an observation with probabilities of the errors satisfying $\epsilon_P + \epsilon_Q \leq 2\epsilon < 1/2$, then

$$\mathcal{H}(p, q) \geq (1 - 4\epsilon) \ln \left(\frac{1}{2\epsilon} \right). \quad (3.6.18)$$

Follow-up: Let p and q be Gaussian densities on \mathbf{R}^k with unit covariance matrices and expectations a, b . Given that $\|b - a\|_2 = 1$, how large should be a sample of independent realizations (drawn either all from p , or all from q) in order to distinguish between P and Q with total error 2.e-6? Give a lower bound on the sample size based on (3.6.18) and its true minimum size (to find it, use the result of the Follow-up in item 1; note that a sample of vectors drawn independently from Gaussian distribution is itself a large Gaussian vector).

3. Hellinger affinity. The *Hellinger affinity* of P and Q is defined as

$$\text{Hel}(p, q) = \int_{\Omega} \sqrt{p(\omega)q(\omega)} d\mu(\omega) \quad \left[= \sum_{\omega \in \Omega} \sqrt{p(\omega)q(\omega)} \text{ in the discrete case} \right]$$

Task 3:

- (a) Prove that the Hellinger affinity is nonnegative, is concave in $(p, q) \geq 0$, does not exceed 1 when p, q are probability densities (and is equal to one if and only if $p = q$), and possesses the following two properties:

$$\text{Hel}(p, q) \leq \sqrt{2 \int \min[p(\omega), q(\omega)] d\mu(\omega)}$$

and

$$\text{Hel}(p^1 \times \dots \times p^k, q^1 \times \dots \times q^k) = \text{Hel}(p^1, q^1) \cdot \dots \cdot \text{Hel}(p^k, q^k).$$

- (b) Derive from the previous item that the total error 2ϵ in distinguishing two hypotheses on distribution of $\omega^k = (\omega_1, \dots, \omega_k) \in \Omega \times \dots \times \Omega$, the first stating that the density of ω^k is $p^{\otimes k}$, and the second stating that this density is $q^{\otimes k}$, admits the lower bound

$$4\epsilon \geq (\text{Hel}(p, q))^{2k}.$$

Follow-up: Compute the Hellinger affinity of two Gaussian densities on \mathbf{R}^k , both with covariance matrices I_k , the means of the densities being a and b . Use this result to derive a lower bound on the sample size considered in the previous Follow-up.

4. Experiment.

Task 4: Carry out the experiment as follows:

- (a) Use the scheme represented in Proposition 3.6.1 to reproduce the results presented in Illustration for the hypotheses B and C. Use the following data on the returns:

$$n = 15, r_i = 1 + \mu_i + \sigma_i \zeta_i, \quad \mu_i = 0.001 + 0.9 \cdot \frac{i-1}{n-1}, \quad \sigma_i = \left[0.9 + 0.2 \cdot \frac{i-1}{n-1} \right] \mu_i, \quad 1 \leq i \leq n,$$

where ζ_i are zero mean random perturbations supported on $[-1, 1]$.

- (b) Find numerically the probability distribution p of $\zeta \in B = [-1, 1]^{15}$ for which the probability of “crisis” $\zeta = -1, 1 \leq i \leq 15$, is as large as possible under the restrictions that
- p is supported on the set of 2^{15} vertices of the box B (i.e., the factors ζ_i take values ± 1 only);
 - the marginal distributions of ζ_i induced by p are the uniform distributions on $\{-1; 1\}$ (i.e., every ζ_i takes values ± 1 with probabilities $1/2$);
 - the covariance matrix of ζ is I_{15} .
- (c) After p is found, take its convex combination with the uniform distribution on the vertices of B to get a distribution P_* on the vertices of B for which the probability of the crisis $P_*(\{\zeta = [-1; \dots; -1]\})$ is exactly 0.01.
- (d) Use the Kullback-Leibler and the Hellinger bounds to bound from below the observation time needed to distinguish, with the total error $2 \cdot 0.01$, between two probability distributions on the vertices of B , namely, P_* and the uniform one (the latter corresponds to the case where $\zeta_i, 1 \leq i \leq 15$, are independent and take values ± 1 with probabilities $1/2$).

3.7 Extremal ellipsoids

We already have met, on different occasions, with the notion of an ellipsoid – a set E in \mathbf{R}^n which can be represented as the image of the unit Euclidean ball under an affine mapping:

$$E = \{x = Au + c \mid u^T u \leq 1\} \quad [A \in \mathbf{M}^{n,q}] \quad (\text{Ell})$$

Ellipsoids are very convenient mathematical entities:

- it is easy to specify an ellipsoid – just to point out the corresponding matrix A and vector c ;
- the family of ellipsoids is closed with respect to affine transformations: the image of an ellipsoid under an affine mapping again is an ellipsoid;
- there are many operations, like minimization of a linear form, computation of volume, etc., which are easy to carry out when the set in question is an ellipsoid, and is difficult to carry out for more general convex sets.

By the indicated reasons, ellipsoids play important role in different areas of applied mathematics; in particular, people use ellipsoids to approximate more complicated sets. Just as a simple motivating example, consider a discrete-time linear time invariant controlled system:

$$\begin{aligned} x(t+1) &= Ax(t) + Bu(t), \quad t = 0, 1, \dots \\ x(0) &= 0 \end{aligned}$$

and assume that the control is norm-bounded:

$$\|u(t)\|_2 \leq 1 \quad \forall t.$$

The question is what is the set X_T of all states “reachable in a given time T ”, i.e., the set of all possible values of $x(T)$. We can easily write down the answer:

$$X_T = \{x = Bu_{T-1} + ABu_{T-2} + A^2Bu_{T-3} + \dots + A^{T-1}Bu_0 \mid \|u_t\|_2 \leq 1, t = 0, \dots, T-1\},$$

but this answer is not “explicit”; just to check whether a given vector x belongs to X_T requires to solve a nontrivial conic quadratic problem, the complexity of the problem being the larger the larger is T . In fact the geometry of X_T may be very complicated, so that there is no possibility to get a “tractable” explicit description of the set. This is why in many applications it makes sense to use “simple” – ellipsoidal - approximations of X_T ; as we shall see, approximations of this type can be computed in a recurrent and computationally efficient fashion.

It turns out that the natural framework for different problems of the “best possible” approximation of convex sets by ellipsoids is given by semidefinite programming. In this Section we intend to consider a number of basic problems of this type.

3.7.1 Preliminaries on ellipsoids

According to our definition, an ellipsoid in \mathbf{R}^n is the image of the unit Euclidean ball in certain \mathbf{R}^q under an affine mapping; e.g., for us a segment in \mathbf{R}^{100} is an ellipsoid; indeed, it is the image of one-dimensional Euclidean ball under affine mapping. In contrast to this, in geometry an ellipsoid in \mathbf{R}^n is usually defined as the image of the n -dimensional unit Euclidean ball under an invertible affine mapping, i.e., as the set of the form (Ell) with additional requirements that $q = n$, i.e., that the matrix A is square, and that it is nonsingular. In order to avoid confusion, let us call these “true” ellipsoids *full-dimensional*. Note that a full-dimensional ellipsoid E admits two nice representations:

- First, E can be represented in the form (Ell) with *positive definite symmetric* A :

$$E = \{x = Au + c \mid u^T u \leq 1\} \quad [A \in \mathbf{S}_{++}^n] \tag{3.7.1}$$

Indeed, it is clear that if a matrix A represents, via (Ell), a given ellipsoid E , the matrix AU , U being an orthogonal $n \times n$ matrix, represents E as well. It is known from Linear Algebra that by multiplying a nonsingular square matrix from the right by a properly chosen orthogonal matrix, we get a positive definite symmetric matrix, so that we always can parameterize a full-dimensional ellipsoid by a positive definite symmetric A .

- Second, E can be given by a strictly convex quadratic inequality:

$$E = \{x \mid (x - c)^T D(x - c) \leq 1\} \quad [D \in \mathbf{S}_{++}^n]. \tag{3.7.2}$$

Indeed, one may take $D = A^{-2}$, where A is the matrix from the representation (3.7.1).

Note that the set (3.7.2) makes sense and is convex when the matrix D is positive semidefinite rather than positive definite. When $D \succeq 0$ is not positive definite, the set (3.7.1) is, geometrically, an “elliptic cylinder” – a shift of the direct product of a full-dimensional ellipsoid in the range space of D and the complementary to this range linear subspace – the kernel of D .

In the sequel we deal a lot with volumes of full-dimensional ellipsoids. Since an invertible affine transformation $x \mapsto Ax + b : \mathbf{R}^n \rightarrow \mathbf{R}^n$ multiplies the volumes of n -dimensional domains by $|\text{Det } A|$, the volume of a full-dimensional ellipsoid E given by (3.7.1) is $\kappa_n \text{Det } A$, where κ_n is the volume of the n -dimensional unit Euclidean ball. In order to avoid meaningless constant factors, it makes sense to pass from the usual n -dimensional volume $\text{mes}_n(G)$ of a domain G to its *normalized volume*

$$\text{Vol}(G) = \kappa_n^{-1} \text{mes}_n(G),$$

i.e., to choose, as the unit of volume, the volume of the unit ball rather than the one of the cube with unit edges. From now on, speaking about volumes of n -dimensional domains, we always mean their normalized volume (and omit the word “normalized”). With this convention, the volume of a full-dimensional ellipsoid E given by (3.7.1) is just

$$\text{Vol}(E) = \text{Det } A,$$

while for an ellipsoid given by (3.7.1) the volume is

$$\text{Vol}(E) = [\text{Det } D]^{-1/2}.$$

3.7.2 Outer and inner ellipsoidal approximations

It was already mentioned that our current goal is to realize how to solve basic problems of “the best” ellipsoidal approximation E of a given set S . There are two types of these problems:

- *Outer approximation*, where we are looking for the “smallest” ellipsoid E containing the set S ;
- *Inner approximation*, where we are looking for the “largest” ellipsoid E contained in the set S .

In both these problems, a natural way to say when one ellipsoid is “smaller” than another one is to compare the volumes of the ellipsoids. The main advantage of this viewpoint is that it results in *affine-invariant* constructions: an invertible affine transformation multiplies volumes of all domains by the same constant and therefore preserves ratios of volumes of the domains.

Thus, what we are interested in are the *largest volume ellipsoid(s) contained in a given set S* and the *smallest volume ellipsoid(s) containing a given set S* . In fact these *extremal ellipsoids* are unique, provided that S is a *solid* – a closed and bounded convex set with a nonempty interior, and are not too bad approximations of the set:

Theorem 3.7.1 [Löwner – Fritz John] *Let $S \subset \mathbf{R}^n$ be a solid. Then*

(i) *There exists and is uniquely defined the largest volume full-dimensional ellipsoid E_{in} contained in S . The concentric to E_{in} n times larger (in linear sizes) ellipsoid contains S ; if S is central-symmetric, then already \sqrt{n} times larger than E_{in} concentric to E_{in} ellipsoid contains S .*

(ii) *There exists and is uniquely defined the smallest volume full-dimensional ellipsoid E_{out} containing S . The concentric to E_{out} n times smaller (in linear sizes) ellipsoid is contained in S ; if S is central-symmetric, then already \sqrt{n} times smaller than E_{out} concentric to E_{out} ellipsoid is contained in S .*

The proof is the subject of Exercise 3.38.

The existence of extremal ellipsoids is, of course, a good news; but how to compute these ellipsoids? The possibility to compute efficiently (nearly) extremal ellipsoids heavily depends on the description of S . Let us start with two simple examples.

3.7.2.1 Inner ellipsoidal approximation of a polytope

Let S be a polyhedral set given by a number of linear equalities:

$$S = \{x \in \mathbf{R}^n \mid a_i^T x \leq b_i, i = 1, \dots, m\}.$$

Proposition 3.7.1 *Assume that S is a full-dimensional polytope (i.e., is bounded and possesses a nonempty interior). Then the largest volume ellipsoid contained in S is*

$$E = \{x = Z_* u + z_* \mid u^T u \leq 1\},$$

where Z_*, z_* are given by an optimal solution to the following semidefinite program:

$$\begin{aligned} & \text{maximize} && t \\ & \text{s.t.} && \\ & (a) && t \leq (\text{Det } Z)^{1/n}, \\ & (b) && Z \succeq 0, \\ & (c) && \|Z a_i\|_2 \leq b_i - a_i^T z, \quad i = 1, \dots, m, \end{aligned} \tag{In}$$

with the design variables $Z \in \mathbf{S}^n, z \in \mathbf{R}^n, t \in \mathbf{R}$.

Note that (In) indeed is a semidefinite program: both (In.a) and (In.c) can be represented by LMIs, see Examples 18d and **1-17** in Section 3.2.

Proof. Indeed, an ellipsoid (3.7.1) is contained in S if and only if

$$a_i^T (A u + c) \leq b_i \quad \forall u : u^T u \leq 1,$$

or, which is the same, if and only if

$$\|A a_i\|_2 + a_i^T c = \max_{u: u^T u \leq 1} [a_i^T A u + a_i^T c] \leq b_i.$$

Thus, (In.b – c) just express the fact that the ellipsoid $\{x = Zu + z \mid u^T u \leq 1\}$ is contained in S , so that (In) is nothing but the problem of maximizing (a positive power of) the volume of an ellipsoid over ellipsoids contained in S . ■

We see that if S is a polytope given by a set of linear inequalities, then the problem of the best inner ellipsoidal approximation of S is an explicit semidefinite program and as such can be efficiently solved. In contrast to this, if S is a polytope given as a convex hull of finite set:

$$S = \text{Conv}\{x_1, \dots, x_m\},$$

then the problem of the best inner ellipsoidal approximation of S is “computationally intractable” – in this case, it is difficult just to check whether a given candidate ellipsoid is contained in S .

3.7.2.2 Outer ellipsoidal approximation of a finite set

Let S be a polyhedral set given as a convex hull of a finite set of points:

$$S = \text{Conv}\{x_1, \dots, x_m\}.$$

Proposition 3.7.2 Assume that S is a full-dimensional polytope (i.e., possesses a nonempty interior). Then the smallest volume ellipsoid containing S is

$$E = \{x \mid (x - c_*)^T D_*(x - c_*) \leq 1\},$$

where c_*, D_* are given by an optimal solution (t_*, Z_*, z_*, s_*) to the semidefinite program

$$\begin{aligned} & \text{maximize} && t \\ & \text{s.t.} && \\ & (a) && t \leq (\text{Det}Z)^{1/n}, \\ & (b) && Z \succeq 0, \\ & (c) && \begin{pmatrix} s & z^T \\ z & Z \end{pmatrix} \succeq 0, \\ & (d) && x_i^T Z x_i - 2x_i^T z + s \leq 1, \quad i = 1, \dots, m, \end{aligned} \tag{Out}$$

with the design variables $Z \in \mathbf{S}^n, z \in \mathbf{R}^n, t, s \in \mathbf{R}$ via the relations

$$D_* = Z_*; c_* = Z_*^{-1}z_*.$$

Note that (Out) indeed is a semidefinite program, cf. Proposition 3.7.1.

Proof. Indeed, let us pass in the description (3.7.2) from the “parameters” D, c to the parameters $Z = D, z = Dc$, thus coming to the representation

$$E = \{x \mid x^T Z x - 2x^T z + z^T Z^{-1}z \leq 1\}. \tag{!}$$

The ellipsoid of the latter type contains the points x_1, \dots, x_m if and only if

$$x_i^T Z x_i - 2x_i^T z + z^T Z^{-1}z \leq 1, \quad i = 1, \dots, m,$$

or, which is the same, if and only if there exists $s \geq z^T Z^{-1}z$ such that

$$x_i^T Z x_i - 2x_i^T z + s \leq 1, \quad i = 1, \dots, m.$$

Recalling Lemma on the Schur Complement, we see that the constraints (Out.b-d) say exactly that the ellipsoid (!) contains the points x_1, \dots, x_m . Since the volume of such an ellipsoid is $(\text{Det}Z)^{-1/2}$, (Out) is the problem of maximizing a negative power of the volume of an ellipsoid containing the finite set $\{x_1, \dots, x_m\}$, i.e., the problem of finding the smallest volume ellipsoid containing this finite set. It remains to note that an ellipsoid is convex, so that it is exactly the same – to say that it contains a finite set $\{x_1, \dots, x_m\}$ and to say that it contains the convex hull of this finite set. ■

We see that if S is a polytope given as a convex hull of a finite set, then the problem of the best outer ellipsoidal approximation of S is an explicit semidefinite program and as such can be efficiently solved. In contrast to this, if S is a polytope given by a list of inequality constraints, then the problem of the best outer ellipsoidal approximation of S is “computationally intractable” – in this case, it is difficult just to check whether a given candidate ellipsoid contains S .

3.7.3 Ellipsoidal approximations of unions/intersections of ellipsoids

Speaking informally, Proposition 3.7.1 deals with inner ellipsoidal approximation of the intersection of “degenerate” ellipsoids, namely, half-spaces (a half-space is just a very large Euclidean ball!) Similarly, Proposition 3.7.2 deals with the outer ellipsoidal approximation of the union of degenerate ellipsoids, namely, points (a point is just a ball of zero radius!). We are about to demonstrate that when passing from “degenerate” ellipsoids to the “normal” ones, we still have a possibility to reduce the corresponding approximation problems to explicit semidefinite programs. The key observation here is as follows:

Proposition 3.7.3 [8] *An ellipsoid*

$$E = E(Z, z) \equiv \{x = Zu + z \mid u^T u \leq 1\} \quad [Z \in \mathbf{M}^{n,q}]$$

is contained in the full-dimensional ellipsoid

$$W = W(Y, y) \equiv \{x \mid (x - y)^T Y^T Y (x - y) \leq 1\} \quad [Y \in \mathbf{M}^{n,n}, \text{Det} Y \neq 0]$$

if and only if there exists λ such that

$$\begin{pmatrix} I_n & Y(z-y) & YZ \\ (z-y)^T Y^T & 1-\lambda & \\ Z^T Y^T & & \lambda I_q \end{pmatrix} \succeq 0 \quad (3.7.3)$$

as well as if and only if there exists λ such that

$$\begin{pmatrix} Y^{-1}(Y^{-1})^T & z-y & Z \\ (z-y)^T & 1-\lambda & \\ Z^T & & \lambda I_q \end{pmatrix} \succeq 0 \quad (3.7.4)$$

Proof. We clearly have

$$\begin{aligned} E &\subset W \\ u^T u \leq 1 &\Rightarrow (Zu + z - y)^T Y^T Y (Zu + z - y) \leq 1 \\ u^T u \leq t^2 &\Rightarrow (Zu + t(z-y))^T Y^T Y (Zu + t(z-y)) \leq t^2 \\ \exists \lambda \geq 0 : [t^2 - (Zu + t(z-y))^T Y^T Y (Zu + t(z-y))] - \lambda[t^2 - u^T u] &\geq 0 \quad \forall(u, t) \\ \exists \lambda \geq 0 : \begin{pmatrix} 1-\lambda-(z-y)^T Y^T Y(z-y) & -(z-y)^T Y^T Y Z \\ -Z^T Y^T Y(z-y) & \lambda I_q - Z^T Y^T Y Z \end{pmatrix} &\succeq 0 \\ \exists \lambda \geq 0 : \begin{pmatrix} 1-\lambda & \\ & \lambda I_q \end{pmatrix} - \begin{pmatrix} (z-y)^T Y^T \\ Z^T Y^T \end{pmatrix} \begin{pmatrix} Y(z-y) & YZ \end{pmatrix} &\succeq 0 \end{aligned}$$

Now note that in view of Lemma on the Schur Complement the matrix

$$\begin{pmatrix} 1-\lambda & \\ & \lambda I_q \end{pmatrix} - \begin{pmatrix} (z-y)^T Y^T \\ Z^T Y^T \end{pmatrix} \begin{pmatrix} Y(z-y) & YZ \end{pmatrix}$$

is positive semidefinite if and only if the matrix in (3.7.3) is so. Thus, $E \subset W$ if and only if there exists a nonnegative λ such that the matrix in (3.7.3), let it be called $P(\lambda)$, is positive

semidefinite. Since the latter matrix can be positive semidefinite only when $\lambda \geq 0$, we have proved the first statement of the proposition. To prove the second statement, note that the matrix in (3.7.4), let it be called $Q(\lambda)$, is closely related to $P(\lambda)$:

$$Q(\lambda) = SP(\lambda)S^T, \quad S = \begin{pmatrix} Y^{-1} & & \\ & 1 & \\ & & I_q \end{pmatrix} \succ 0,$$

so that $Q(\lambda)$ is positive semidefinite if and only if $P(\lambda)$ is so. ■

Here are some consequences of Proposition 3.7.3.

3.7.3.1 Inner ellipsoidal approximation of the intersection of full-dimensional ellipsoids

Let

$$W_i = \{x \mid (x - c_i)^T B_i^2 (x - c_i) \leq 1\} \quad [B_i \in \mathbf{S}_{++}^n],$$

$i = 1, \dots, m$, be given full-dimensional ellipsoids in \mathbf{R}^n ; assume that the intersection W of these ellipsoids possesses a nonempty interior. Then the problem of the best inner ellipsoidal approximation of W is the explicit semidefinite program

$$\begin{aligned} \text{maximize} \quad & t \\ \text{s.t.} \quad & \begin{aligned} t &\leq (\text{Det}Z)^{1/n}, \\ \begin{pmatrix} I_n & B_i(z - c_i) & B_i Z \\ (z - c_i)^T B_i & 1 - \lambda_i & \lambda_i I_n \\ Z B_i & & Z \end{pmatrix} &\succeq 0, \quad i = 1, \dots, m, \\ & Z \succeq 0 \end{aligned} \end{aligned} \tag{InEll}$$

with the design variables $Z \in \mathbf{S}^n$, $z \in \mathbf{R}^n$, $\lambda_i, t \in \mathbf{R}$. The largest ellipsoid contained in $W = \bigcap_{i=1}^m W_i$ is given by an optimal solution $Z_*, z_*, t_*, \{\lambda_i^*\}$ of (InEll) via the relation

$$E = \{x = Z_* u + z_* \mid u^T u \leq 1\}.$$

Indeed, by Proposition 3.7.3 the LMIs

$$\begin{pmatrix} I_n & B_i(z - c_i) & B_i Z \\ (z - c_i)^T B_i & 1 - \lambda_i & \lambda_i I_n \\ Z B_i & & Z \end{pmatrix} \succeq 0, \quad i = 1, \dots, m$$

express the fact that the ellipsoid $\{x = Zu + z \mid u^T u \leq 1\}$ with $Z \succeq 0$ is contained in every one of the ellipsoids W_i , i.e., is contained in the intersection W of these ellipsoids. Consequently, (InEll) is exactly the problem of maximizing (a positive power of) the volume of an ellipsoid over the ellipsoids contained in W .

3.7.3.2 Outer ellipsoidal approximation of the union of ellipsoids

Let

$$W_i = \{x = A_i u + c_i \mid u^T u \leq 1\} \quad [A_i \in \mathbf{M}^{n, k_i}],$$

$i = 1, \dots, m$, be given ellipsoids in \mathbf{R}^n ; assume that the convex hull W of the union of these ellipsoids possesses a nonempty interior. Then the problem of the best outer ellipsoidal approximation of W is the explicit semidefinite program

$$\begin{aligned} & \text{maximize} && t \\ & \text{s.t.} && \\ & && t \leq (\text{Det}Y)^{1/n}, \\ & \begin{pmatrix} I_n & Yc_i - z & YA_i \\ (Yc_i - z)^T & 1 - \lambda_i & \lambda_i I_{k_i} \\ A_i^T Y & & \end{pmatrix} \succeq 0, \quad i = 1, \dots, m, \\ & && Y \succeq 0 \end{aligned} \tag{OutEll}$$

with the design variables $Y \in \mathbf{S}^n$, $z \in \mathbf{R}^n$, $\lambda_i, t \in \mathbf{R}$. The smallest ellipsoid containing $W = \text{Conv}(\bigcup_{i=1}^m W_i)$ is given by an optimal solution $(Y_*, z_*, t_*, \{\lambda_i^*\})$ of (OutEll) via the relation

$$E = \{x \mid (x - y_*)Y_*^2(x - y_*) \leq 1\}, \quad y_* = Y_*^{-1}z_*.$$

Indeed, by Proposition 3.7.3 for $Y \succ 0$ the LMIs

$$\begin{pmatrix} I_n & Yc_i - z & YA_i \\ (Yc_i - z)^T & 1 - \lambda_i & \lambda_i I_{k_i} \\ A_i^T Y & & \end{pmatrix} \succeq 0, \quad i = 1, \dots, m$$

express the fact that the ellipsoid $E = \{x \mid (x - Y^{-1}z)^T Y^2(x - Y^{-1}z) \leq 1\}$ contains every one of the ellipsoids W_i , i.e., contains the convex hull W of the union of these ellipsoids. The volume of the ellipsoid E is $(\text{Det}Y)^{-1}$; consequently, (OutEll) is exactly the problem of maximizing a *negative* power (i.e., of minimizing a positive power) of the volume of an ellipsoid over the ellipsoids containing W .

3.7.4 Approximating sums of ellipsoids

Let us come back to our motivating example, where we were interested to build ellipsoidal approximation of the set X_T of all states $x(T)$ where a given discrete time invariant linear system

$$\begin{aligned} x(t+1) &= Ax(t) + Bu(t), \quad t = 0, \dots, T-1 \\ x(0) &= 0 \end{aligned}$$

can be driven in time T by a control $u(\cdot)$ satisfying the norm bound

$$\|u(t)\|_2 \leq 1, \quad t = 0, \dots, T-1.$$

How could we build such an approximation recursively? Let X_t be the set of all states where the system can be driven in time $t \leq T$, and assume that we have already built inner and outer ellipsoidal approximations E_{in}^t and E_{out}^t of the set X_t :

$$E_{\text{in}}^t \subset X_t \subset E_{\text{out}}^t.$$

Let also

$$E = \{x = Bu \mid u^T u \leq 1\}.$$

Then the set

$$F_{\text{in}}^{t+1} = AE_{\text{in}}^t + E \equiv \{x = Ay + z \mid y \in E_{\text{in}}^t, z \in E\}$$

clearly is contained in X_{t+1} , so that a natural recurrent way to define an inner ellipsoidal approximation of X_{t+1} is to take as E_{in}^{t+1} the largest volume ellipsoid contained in F_{in}^{t+1} . Similarly, the set

$$F_{\text{out}}^{t+1} = AE_{\text{out}}^t + E \equiv \{x = Ay + z \mid y \in E_{\text{out}}^t, z \in E\}$$

clearly covers X_{t+1} , and the natural recurrent way to define an outer ellipsoidal approximation of X_{t+1} is to take as E_{out}^{t+1} the smallest volume ellipsoid containing F_{out}^{t+1} .

Note that the sets F_{in}^{t+1} and F_{out}^{t+1} are of the same structure: each of them is the *arithmetic sum* $\{x = v + w \mid v \in V, w \in W\}$ of two ellipsoids V and W . Thus, we come to the problem as follows: *Given two ellipsoids W, V , find the best inner and outer ellipsoidal approximations of their arithmetic sum $W + V$.* In fact, it makes sense to consider a little bit more general problem:

Given m ellipsoids W_1, \dots, W_m in \mathbf{R}^n , find the best inner and outer ellipsoidal approximations of the arithmetic sum

$$W = \{x = w_1 + w_2 + \dots + w_m \mid w_i \in W_i, i = 1, \dots, m\}$$

of the ellipsoids W_1, \dots, W_m .

In fact, we have posed two different problems: the one of inner approximation of W (let this problem be called (I)) and the other one, let it be called (O), of outer approximation. It seems that *in general* both these problems are difficult (at least when m is not once for ever fixed). There exist, however, “computationally tractable” approximations of both (I) and (O) we are about to consider.

In considerations to follow we assume, for the sake of simplicity, that the ellipsoids W_1, \dots, W_m are full-dimensional (which is not a severe restriction – a “flat” ellipsoid can be easily approximated by a “nearly flat” full-dimensional ellipsoid). Besides this, we may assume without loss of generality that all our ellipsoids W_i are centered at the origin. Indeed, we have $W_i = c_i + V_i$, where c_i is the center of W_i and $V_i = W_i - c_i$ is centered at the origin; consequently,

$$W_1 + \dots + W_m = (c_1 + \dots + c_m) + (V_1 + \dots + V_m),$$

so that the problems (I) and (O) for the ellipsoids W_1, \dots, W_m can be straightforwardly reduced to similar problems for the centered at the origin ellipsoids V_1, \dots, V_m .

3.7.4.1 Problem (O)

Let the ellipsoids W_1, \dots, W_m be represented as

$$W_i = \{x \in \mathbf{R}^n \mid x^T B_i x \leq 1\} \quad [B_i \succ 0].$$

Our strategy to approximate (O) is very natural: we intend to build a parametric family of ellipsoids in such a way that, first, every ellipsoid from the family contains the arithmetic sum $W_1 + \dots + W_m$ of given ellipsoids, and, second, the problem of finding the smallest volume ellipsoid *within the family* is a “computationally tractable” problem (specifically, is an explicit

semidefinite program)²³⁾. The seemingly simplest way to build the desired family was proposed in [8] and is based on the idea of semidefinite relaxation. Let us start with the observation that an ellipsoid

$$W[Z] = \{x \mid x^T Z x \leq 1\} \quad [Z \succ 0]$$

contains $W_1 + \dots + W_m$ if and only if the following implication holds:

$$\left\{ \{x^i \in \mathbf{R}^n\}_{i=1}^m, [x^i]^T B_i x^i \leq 1, i = 1, \dots, m \right\} \Rightarrow (x^1 + \dots + x^m)^T Z (x^1 + \dots + x^m) \leq 1. \quad (*)$$

Now let B^i be $(nm) \times (nm)$ block-diagonal matrix with m diagonal blocks of the size $n \times n$ each, such that all diagonal blocks, except the i -th one, are zero, and the i -th block is the $n \times n$ matrix B_i . Let also $M[Z]$ denote $(mn) \times (mn)$ block matrix with m^2 blocks of the size $n \times n$ each, every of these blocks being the matrix Z . This is how B^i and $M[Z]$ look in the case of $m = 2$:

$$B^1 = \begin{bmatrix} B_1 & \\ & \end{bmatrix}, \quad B^2 = \begin{bmatrix} & \\ & B_2 \end{bmatrix}, \quad M[Z] = \begin{bmatrix} Z & Z \\ Z & Z \end{bmatrix}.$$

Validity of implication $(*)$ clearly is equivalent to the following fact:

(*.1) For every (mn) -dimensional vector x such that

$$x^T B^i x \equiv \text{Tr}(B^i \underbrace{x x^T}_{X[x]}) \leq 1, \quad i = 1, \dots, m,$$

one has

$$x^T M[Z] x \equiv \text{Tr}(M[Z] X[x]) \leq 1.$$

Now we can use the standard trick: the rank one matrix $X[x]$ is positive semidefinite, so that we for sure enforce the validity of the above fact when enforcing the following stronger fact:

(*.2) For every $(mn) \times (mn)$ symmetric positive semidefinite matrix X such that

$$\text{Tr}(B^i X) \leq 1, \quad i = 1, \dots, m,$$

one has

$$\text{Tr}(M[Z] X) \leq 1.$$

We have arrived at the following result.

(D) Let a positive definite $n \times n$ matrix Z be such that the optimal value in the semidefinite program

$$\max_X \left\{ \text{Tr}(M[Z] X) \mid \text{Tr}(B^i X) \leq 1, \quad i = 1, \dots, m, \quad X \succeq 0 \right\} \quad (\text{SDP})$$

is ≤ 1 . Then the ellipsoid

$$W[Z] = \{x \mid x^T Z x \leq 1\}$$

contains the arithmetic sum $W_1 + \dots + W_m$ of the ellipsoids $W_i = \{x \mid x^T B_i x \leq 1\}$.

²³⁾ Note that we, in general, do not pretend that our parametric family includes all ellipsoids containing $W_1 + \dots + W_m$, so that the ellipsoid we end with should be treated as nothing more than a “computable surrogate” of the smallest volume ellipsoid containing the sum of W_i 's.

We are basically done: the set of those symmetric matrices Z for which the optimal value in (SDP) is ≤ 1 is SD-representable; indeed, the problem is clearly strictly feasible, and Z affects, in a linear fashion, the objective of the problem only. On the other hand, the optimal value in a strictly feasible semidefinite *maximization* program is a SDr function of the objective (“semidefinite version” of Proposition 2.4.4). Consequently, the set of those Z for which the optimal value in (SDP) is ≤ 1 is SDr (as the inverse image, under affine mapping, of the level set of a SDr function). Thus, the “parameter” Z of those ellipsoids $W[Z]$ which satisfy the premise in **(D)** and thus contain $W_1 + \dots + W_m$ varies in an SDr set \mathcal{Z} . Consequently, the problem of finding the smallest volume ellipsoid in the family $\{W[Z]\}_{Z \in \mathcal{Z}}$ is equivalent to the problem of maximizing a positive power of $\text{Det}(Z)$ over the SDr set \mathcal{Z} , i.e., is equivalent to a semidefinite program.

It remains to build the aforementioned semidefinite program. By the Conic Duality Theorem the optimal value in the (clearly strictly feasible) maximization program (SDP) is ≤ 1 if and only if the dual problem

$$\min_{\lambda} \left\{ \sum_{i=1}^m \lambda_i \mid \sum_i \lambda_i B^i \succeq M[Z], \lambda_i \geq 0, i = 1, \dots, m \right\}.$$

admits a feasible solution with the value of the objective ≤ 1 , or, which is clearly the same (why?), admits a feasible solution with the value of the objective equal 1. In other words, whenever $Z \succeq 0$ is such that $M[Z]$ is \preceq a convex combination of the matrices B^i , the set

$$W[Z] = \{x \mid x^T Z x \leq 1\}$$

(which is an ellipsoid when $Z \succ 0$) contains the set $W_1 + \dots + W_m$. We have arrived at the following result (see [8], Section 3.7.4):

Proposition 3.7.4 *Given m centered at the origin full-dimensional ellipsoids*

$$W_i = \{x \in \mathbf{R}^n \mid x^T B_i x \leq 1\} \quad [B_i \succ 0],$$

$i = 1, \dots, m$, in \mathbf{R}^n , let us associate with these ellipsoids the semidefinite program

$$\max_{t, Z, \lambda} \left\{ t \mid \begin{array}{l} t \leq \text{Det}^{1/n}(Z) \\ \sum_{i=1}^m \lambda_i B^i \succeq M[Z] \\ \lambda_i \geq 0, i = 1, \dots, m \\ Z \succeq 0 \\ \sum_{i=1}^m \lambda_i = 1 \end{array} \right\} \quad (\tilde{\mathcal{O}})$$

where B^i is the $(mn) \times (mn)$ block-diagonal matrix with blocks of the size $n \times n$ and the only nonzero diagonal block (the i -th one) equal to B_i , and $M[Z]$ is the $(mn) \times (mn)$ matrix partitioned into m^2 blocks, every one of them being Z . Every feasible solution (Z, \dots) to this program with positive value of the objective produces ellipsoid

$$W[Z] = \{x \mid x^T Z x \leq 1\}$$

which contains $W_1 + \dots + W_m$, and the volume of this ellipsoid is at most $t^{-n/2}$. The smallest volume ellipsoid which can be obtained in this way is given by (any) optimal solution of $(\tilde{\mathcal{O}})$.

How “conservative” is (\tilde{O}) ? The ellipsoid $W[Z^*]$ given by the optimal solution of (\tilde{O}) contains the arithmetic sum W of the ellipsoids W_i , but not necessarily is the smallest volume ellipsoid containing W ; all we know is that this ellipsoid is the smallest volume one in certain subfamily of the family of all ellipsoids containing W . “In the nature” there exists the “true” smallest volume ellipsoid $W[Z^{**}] = \{x \mid x^T Z^{**} x \leq 1\}$, $Z^{**} \succ 0$, containing W . It is natural to ask how large could be the ratio

$$\vartheta = \frac{\text{Vol}(W[Z^*])}{\text{Vol}(W[Z^{**}])}.$$

The answer is as follows:

Proposition 3.7.5 *One has $\vartheta \leq (\frac{\pi}{2})^{n/2}$.*

Note that the bound stated by Proposition 3.7.5 is not as bad as it looks: the natural way to compare the “sizes” of two n -dimensional bodies E', E'' is to look at the ratio of their average linear sizes $\left(\frac{\text{Vol}(E')}{\text{Vol}(E'')}\right)^{1/n}$ (it is natural to assume that shrinking a body by certain factor, say, 2, we reduce the “size” of the body exactly by this factor, and not by 2^n). With this approach, the “level of non-optimality” of $W[Z^*]$ is no more than $\sqrt{\pi/2} = 1.253\dots$, i.e., is within 25% margin.

Proof of Proposition 3.7.5: Since Z^{**} contains W , the implication $(*)$ holds true, i.e., one has

$$\max_{x \in \mathbf{R}^{mn}} \{x^T M[Z^{**}] x \mid x^T B^i x \leq 1, i = 1, \dots, m\} \leq 1.$$

Since the matrices B^i , $i = 1, \dots, m$, commute and $M[Z^{**}] \succeq 0$, we can apply Proposition 3.8.1 (see Section 3.8.5) to conclude that there exist nonnegative μ_i , $i = 1, \dots, m$, such that

$$M[Z^{**}] \preceq \sum_{i=1}^m \mu_i B^i, \quad \sum_i \mu_i \leq \frac{\pi}{2}.$$

It follows that setting $\lambda_i = \left(\sum_j \mu_j\right)^{-1} \mu_i$, $Z = \left(\sum_j \mu_j\right)^{-1} Z^{**}$, $t = \text{Det}^{1/n}(Z)$, we get a feasible solution of (\tilde{O}) . Recalling the origin of Z^* , we come to

$$\text{Vol}(W[Z^*]) \leq \text{Vol}(W[Z]) = \left(\sum_j \mu_j\right)^{n/2} \text{Vol}(W[Z^{**}]) \leq (\pi/2)^{n/2} \text{Vol}(W[Z^{**}]),$$

as claimed. ■

Problem (O), the case of “co-axial” ellipsoids. Consider the *co-axial case* – the one when there exist coordinates (not necessarily orthogonal) such that all m quadratic forms defining the ellipsoids W_i are diagonal in these coordinates, or, which is the same, there exists a nonsingular matrix C such that all the matrices $C^T B_i C$, $i = 1, \dots, m$, are diagonal. Note that the case of $m = 2$ always is co-axial – Linear Algebra says that every two homogeneous quadratic forms, at least one of the forms being positive outside of the origin, become diagonal in a properly chosen coordinates.

We are about to prove that

(E) *In the “co-axial” case, (\tilde{O}) yields the smallest in volume ellipsoid containing $W_1 + \dots + W_m$.*

Consider the co-axial case. Since we are interested in volume-related issues, and the ratio of volumes remains unchanged under affine transformations, we can assume w.l.o.g. that the matrices B_i defining the ellipsoids $W_i = \{x \mid x^T B_i x \leq 1\}$ are positive definite and diagonal; let b_ℓ^i be the ℓ -th diagonal entry of B_i , $\ell = 1, \dots, n$.

By the Fritz John Theorem, “in the nature” there exists a unique smallest volume ellipsoid W_* which contains $W_1 + \dots + W_m$; from uniqueness combined with the fact that the sum of our ellipsoids is symmetric w.r.t. the origin it follows that this optimal ellipsoid W_* is centered at the origin:

$$W_* = \{x \mid x^T Z_* x \leq 1\}$$

with certain positive definite matrix Z_* .

Our next observation is that the matrix Z_* is diagonal. Indeed, let E be a diagonal matrix with diagonal entries ± 1 . Since all B_i 's are diagonal, the sum $W_1 + \dots + W_m$ remains invariant under multiplication by E :

$$x \in W_1 + \dots + W_m \Leftrightarrow Ex \in W_1 + \dots + W_m.$$

It follows that the ellipsoid $E(W_*) = \{x \mid x^T (E^T Z_* E) x \leq 1\}$ covers $W_1 + \dots + W_m$ along with W_* and of course has the same volume as W_* ; from the uniqueness of the optimal ellipsoid it follows that $E(W_*) = W_*$, whence $E^T Z_* E = Z_*$ (why?). Since the concluding relation should be valid for all diagonal matrices E with diagonal entries ± 1 , Z_* must be diagonal.

Now assume that the set

$$W(z) = \{x \mid x^T \text{Diag}(z) x \leq 1\} \quad (3.7.5)$$

given by a nonnegative vector z contains $W_1 + \dots + W_m$. Then the following implication holds true:

$$\forall \{x_\ell^i\}_{\substack{i=1, \dots, m \\ \ell=1, \dots, n}} : \sum_{\ell=1}^n b_\ell^i (x_\ell^i)^2 \leq 1, i = 1, \dots, m \Rightarrow \sum_{\ell=1}^n z_\ell (x_\ell^1 + x_\ell^2 + \dots + x_\ell^m)^2 \leq 1. \quad (3.7.6)$$

Denoting $y_\ell^i = (x_\ell^i)^2$ and taking into account that $z_\ell \geq 0$, we see that the validity of (3.7.6) implies the validity of the implication

$$\forall \{y_\ell^i \geq 0\}_{\substack{i=1, \dots, m \\ \ell=1, \dots, n}} : \sum_{\ell=1}^n b_\ell^i y_\ell^i \leq 1, i = 1, \dots, m \Rightarrow \sum_{\ell=1}^n z_\ell \left(\sum_{i=1}^m y_\ell^i + 2 \sum_{1 \leq i < j \leq m} \sqrt{y_\ell^i y_\ell^j} \right) \leq 1. \quad (3.7.7)$$

Now let Y be an $(mn) \times (mn)$ symmetric matrix satisfying the relations

$$Y \succeq 0; \text{Tr}(Y B^i) \leq 1, i = 1, \dots, m. \quad (3.7.8)$$

Let us partition Y into m^2 square blocks, and let Y_ℓ^{ij} be the ℓ -th diagonal entry of the ij -th block of Y . For all i, j with $1 \leq i < j \leq m$, and all ℓ , $1 \leq \ell \leq n$, the 2×2 matrix $\begin{pmatrix} Y_\ell^{ii} & Y_\ell^{ij} \\ Y_\ell^{ij} & Y_\ell^{jj} \end{pmatrix}$ is a principal submatrix of Y and therefore is positive semidefinite along with Y , whence

$$Y_\ell^{ij} \leq \sqrt{Y_\ell^{ii} Y_\ell^{jj}}. \quad (3.7.9)$$

In view of (3.7.8), the numbers $y_\ell^i \equiv Y_\ell^{ii}$ satisfy the premise in the implication (3.7.7), so that

$$\begin{aligned} 1 &\geq \sum_{\ell=1}^n z_\ell \left[\sum_{i=1}^m Y_\ell^{ii} + 2 \sum_{1 \leq i < j \leq m} \sqrt{Y_\ell^{ii} Y_\ell^{jj}} \right] && [\text{by (3.7.7)}] \\ &\geq \sum_{\ell=1}^n z_\ell \left[\sum_{i=1}^m Y_\ell^{ii} + 2 \sum_{1 \leq i < j \leq m} Y_\ell^{ij} \right] && [\text{since } z \geq 0 \text{ and by (3.7.9)}] \\ &= \text{Tr}(YM[\text{Diag}(z)]). \end{aligned}$$

Thus, (3.7.8) implies the inequality $\text{Tr}(YM[\text{Diag}(z)]) \leq 1$, i.e., the implication

$$Y \succeq 0, \text{Tr}(YB^i) \leq 1, i = 1, \dots, m \Rightarrow \text{Tr}(YM[\text{Diag}(z)]) \leq 1$$

holds true. Since the premise in this implication is strictly feasible, the validity of the implication, by Semidefinite Duality, implies the existence of nonnegative λ_i , $\sum_i \lambda_i \leq 1$, such that

$$M[\text{Diag}(z)] \preceq \sum_i \lambda_i B^i.$$

Combining our observations, we come to the conclusion as follows:

In the case of diagonal matrices B_i , if the set (3.7.5), given by a nonnegative vector z , contains $W_1 + \dots + W_m$, then the matrix $\text{Diag}(z)$ can be extended to a feasible solution of the problem (\tilde{O}) . Consequently, in the case in question the approximation scheme given by (\tilde{O}) yields the minimum volume ellipsoid containing $W_1 + \dots + W_m$ (since the latter ellipsoid, as we have seen, is of the form (3.7.5) with $z \geq 0$).

It remains to note that the approximation scheme associated with (\tilde{O}) is affine-invariant, so that the above conclusion remains valid when we replace in its premise “the case of diagonal matrices B_i ” with “the co-axial case”.

Remark 3.7.1 In fact, **(E)** is an immediate consequence of the following fact (which, essentially, is proved in the above reasoning):

Let A_1, \dots, A_m, B be symmetric matrices such that the off-diagonal entries of all A_i 's are nonpositive, and the off-diagonal entries of B are nonnegative. Assume also that the system of inequalities

$$x^T A_i x \leq a_i, \quad i = 1, \dots, m \tag{S}$$

is strictly feasible. Then the inequality

$$x^T B x \leq b$$

is a consequence of the system (S) if and only if it is a “linear consequence” of (S), i.e., if and only if there exist nonnegative weights λ_i such that

$$B \preceq \sum_i \lambda_i A_i, \quad \sum_i \lambda_i a_i \leq b.$$

In other words, in the case in question the optimization program

$$\max_x \left\{ x^T B x \mid x^T A_i x \leq a_i, i = 1, \dots, m \right\}$$

and its standard semidefinite relaxation

$$\max_X \left\{ \text{Tr}(BX) \mid X \succeq 0, \text{Tr}(A_i X) \leq a_i, i = 1, \dots, m \right\}$$

share the same optimal value.

3.7.4.2 Problem (I)

Let us represent the given centered at the origin ellipsoids W_i as

$$W_i = \{x \mid x = A_i u \mid u^T u \leq 1\} \quad [\text{Det}(A_i) \neq 0].$$

We start from the following observation:

(F) An ellipsoid $E[Z] = \{x = Z u \mid u^T u \leq 1\}$ ($\text{Det}(Z) \neq 0$) is contained in the sum $W_1 + \dots + W_m$ of the ellipsoids W_i if and only if one has

$$\forall x : \|Z^T x\|_2 \leq \sum_{i=1}^m \|A_i^T x\|_2. \quad (3.7.10)$$

Indeed, assume, first, that there exists a vector x_* such that the inequality in (3.7.10) is violated at $x = x_*$, and let us prove that in this case $E[Z]$ is not contained in the set $W = W_1 + \dots + W_m$. We have

$$\max_{x \in W_i} x^T x = \max [x_*^T A_i u \mid u^T u \leq 1] = \|A_i^T x_*\|_2, \quad i = 1, \dots, m,$$

and similarly

$$\max_{x \in E[Z]} x^T x = \|Z^T x_*\|_2,$$

whence

$$\begin{aligned} \max_{x \in W} x^T x &= \max_{x^i \in W_i} x_*^T (x^1 + \dots + x^m) = \sum_{i=1}^m \max_{x^i \in W_i} x_*^T x^i \\ &= \sum_{i=1}^m \|A_i^T x_*\|_2 < \|Z^T x_*\|_2 = \max_{x \in E[Z]} x^T x, \end{aligned}$$

and we see that $E[Z]$ cannot be contained in W . Vice versa, assume that $E[Z]$ is not contained in W , and let $y \in E[Z] \setminus W$. Since W is a convex compact set and $y \notin W$, there exists a vector x_* such that $x_*^T y > \max_{x \in W} x^T x$, whence, due to the previous computation,

$$\|Z^T x_*\|_2 = \max_{x \in E[Z]} x^T x \geq x_*^T y > \max_{x \in W} x^T x = \sum_{i=1}^m \|A_i^T x_*\|_2,$$

and we have found a point $x = x_*$ at which the inequality in (3.7.10) is violated. Thus, $E[Z]$ is not contained in W if and only if (3.7.10) is not true, which is exactly what should be proved.

A natural way to generate ellipsoids satisfying (3.7.10) is to note that whenever X_i are $n \times n$ matrices of spectral norms

$$|X_i| \equiv \sqrt{\lambda_{\max}(X_i^T X_i)} = \sqrt{\lambda_{\max}(X_i X_i^T)} = \max_x \{\|X_i x\|_2 \mid \|x\|_2 \leq 1\}$$

not exceeding 1, the matrix

$$Z = Z(X_1, \dots, X_m) = A_1 X_1 + A_2 X_2 + \dots + A_m X_m$$

satisfies (3.7.10):

$$\|Z^T x\|_2 = \|[X_1^T A_1^T + \dots + X_m^T A_m^T] x\|_2 \leq \sum_{i=1}^m \|X_i^T A_i^T x\|_2 \leq \sum_{i=1}^m |X_i^T| \|A_i^T x\|_2 \leq \sum_{i=1}^m \|A_i^T x\|_2.$$

Thus, every collection of square matrices X_i with spectral norms not exceeding 1 produces an ellipsoid satisfying (3.7.10) and thus contained in W , and we could use the largest volume ellipsoid of this form (i.e., the one corresponding to the largest $|\text{Det}(A_1X_1 + \dots + A_mX_m)|$) as a surrogate of the largest volume ellipsoid contained in W . Recall that we know how to express a bound on the spectral norm of a matrix via LMI:

$$|X| \leq t \Leftrightarrow \begin{pmatrix} tI_n & -X^T \\ -X & tI_n \end{pmatrix} \succeq 0 \quad [X \in \mathbf{M}^{n,n}]$$

(item 16 of Section 3.2). The difficulty, however, is that the matrix $\sum_{i=1}^m A_i X_i$ specifying the ellipsoid $E(X_1, \dots, X_m)$, although being linear in the “design variables” X_i , is not necessarily symmetric positive semidefinite, and we do not know how to maximize the determinant over general-type square matrices. We may, however, use the following fact from Linear Algebra:

Lemma 3.7.1 *Let $Y = S + C$ be a square matrix represented as the sum of a symmetric matrix S and a skew-symmetric (i.e., $C^T = -C$) matrix C . Assume that S is positive definite. Then*

$$|\text{Det}(Y)| \geq \text{Det}(S).$$

Proof. We have $Y = S + C = S^{1/2}(I + \Sigma)S^{1/2}$, where $\Sigma = S^{-1/2}CS^{-1/2}$ is skew-symmetric along with C . We have $|\text{Det}(Y)| = \text{Det}(S)|\text{Det}(I + \Sigma)|$; it remains to note that all eigenvalues of the skew-symmetric matrix Σ are purely imaginary, so that the eigenvalues of $I + \Sigma$ are ≥ 1 in absolute value, whence $|\text{Det}(I + \Sigma)| \geq 1$. ■

In view of Lemma, it makes sense to impose on X_1, \dots, X_m , besides the requirement that their spectral norms are ≤ 1 , also the requirement that the “symmetric part”

$$S(X_1, \dots, X_m) = \frac{1}{2} \left[\sum_{i=1}^m A_i X_i + \sum_{i=1}^m X_i^T A_i \right]$$

of the matrix $\sum_i A_i X_i$ is positive semidefinite, and to maximize under these constraints the quantity $\text{Det}(S(X_1, \dots, X_m))$ – a lower bound on the volume of the ellipsoid $E(X_1, \dots, X_m)$. With this approach, we come to the following result:

Proposition 3.7.6 *Let $W_i = \{x = A_i u \mid u^T u \leq 1\}$, $A_i \succ 0$, $i = 1, \dots, m$. Consider the semidefinite program*

$$\begin{aligned} & \text{maximize} && t \\ & \text{s.t.} && \\ (a) & t \leq \left(\text{Det} \left(\frac{1}{2} \sum_{i=1}^m [X_i^T A_i + A_i X_i] \right) \right)^{1/n} \\ (b) & \sum_{i=1}^m [X_i^T A_i + A_i X_i] \succeq 0 \\ (c) & \begin{pmatrix} I_n & -X_i^T \\ -X_i & I_n \end{pmatrix} \succeq 0, \quad i = 1, \dots, m \end{aligned} \tag{I}$$

with design variables $X_1, \dots, X_m \in \mathbf{M}^{n,n}$, $t \in \mathbf{R}$. Every feasible solution $(\{X_i\}, t)$ to this problem produces the ellipsoid

$$E(X_1, \dots, X_m) = \{x = (\sum_{i=1}^m A_i X_i)u \mid u^T u \leq 1\}$$

contained in the arithmetic sum $W_1 + \dots + W_m$ of the original ellipsoids, and the volume of this ellipsoid is at least t^n . The largest volume ellipsoid which can be obtained in this way is associated with (any) optimal solution to (\tilde{I}) .

In fact, problem (I) is equivalent to the problem

$$|\text{Det}(\sum_{i=1}^m A_i X_i)| \rightarrow \max \mid |X_i| \leq 1, i = 1, \dots, m \quad (3.7.11)$$

we have started with, since the latter problem always has an optimal solution $\{X_i^*\}$ with positive semidefinite symmetric matrix $G_* = \sum_{i=1}^m A_i X_i^*$. Indeed, let $\{X_i^+\}$ be an optimal solution of the problem. The matrix $G_+ = \sum_{i=1}^m A_i X_i^+$, as every $n \times n$ square matrix, admits a representation $G_+ = G_* U$, where G_+ is a positive semidefinite symmetric, and U is an orthogonal matrix. Setting $X_i^* = X_i U^T$, we convert $\{X_i^+\}$ into a new feasible solution of (3.7.11); for this solution $\sum_{i=1}^m A_i X_i^* = G_* \succeq 0$, and $\text{Det}(G_+) = \text{Det}(G_*)$, so that the new solution is optimal along with $\{X_i^+\}$.

Problem (I), the co-axial case. We are about to demonstrate that in the co-axial case, when in properly chosen coordinates in \mathbf{R}^n the ellipsoids W_i can be represented as

$$W_i = \{x = A_i u \mid u^T u \leq 1\}$$

with positive definite diagonal matrices A_i , the above scheme yields the best (the largest volume) ellipsoid among those contained in $W = W_1 + \dots + W_m$. Moreover, this ellipsoid can be pointed out explicitly – it is exactly the ellipsoid $E[Z]$ with $Z = Z(I_n, \dots, I_n) = A_1 + \dots + A_m$!

The announced fact is nearly evident. Assuming that A_i are positive definite and diagonal, consider the parallelotope

$$\widehat{W} = \{x \in \mathbf{R}^n \mid |x_j| \leq \ell_j = \sum_{i=1}^m [A_i]_{jj}, j = 1, \dots, n\}.$$

This parallelotope clearly contains W (why?), and the largest volume ellipsoid contained in \widehat{W} clearly is the ellipsoid

$$\{x \mid \sum_{j=1}^n \ell_j^{-2} x_j^2 \leq 1\},$$

i.e., is nothing else but the ellipsoid $E[A_1 + \dots + A_m]$. As we know from our previous considerations, the latter ellipsoid is contained in W , and since it is the largest volume ellipsoid among those contained in the set $\widehat{W} \supset W$, it is the largest volume ellipsoid contained in W as well.

Example. In the example to follow we are interested to understand what is the domain D_T on the 2D plane which can be reached by a trajectory of the differential equation

$$\frac{d}{dt} \begin{pmatrix} x_1(t) \\ x_2(t) \end{pmatrix} = \underbrace{\begin{pmatrix} -0.8147 & -0.4163 \\ 0.8167 & -0.1853 \end{pmatrix}}_A \begin{pmatrix} x_1(t) \\ x_2(t) \end{pmatrix} + \begin{pmatrix} u_1(t) \\ 0.7071 u_2(t) \end{pmatrix}, \quad \begin{pmatrix} x_1(0) \\ x_2(0) \end{pmatrix} = \begin{pmatrix} 0 \\ 0 \end{pmatrix}$$

in T sec under a piecewise-constant control $u(t) = \begin{pmatrix} u_1(t) \\ u_2(t) \end{pmatrix}$ which switches from one constant value to another one every $\Delta t = 0.01$ sec and is subject to the norm bound

$$\|u(t)\|_2 \leq 1 \quad \forall t.$$

The system is stable (the eigenvalues of A are $-0.5 \pm 0.4909i$). In order to build D_T , note that the states of the system at time instants $k\Delta t$, $k = 0, 1, 2, \dots$ are the same as the states $x[k] = \begin{pmatrix} x_1(k\Delta t) \\ x_2(k\Delta t) \end{pmatrix}$ of the discrete time system

$$x[k+1] = \underbrace{\exp\{A\Delta t\}}_S x[k] + \underbrace{\left[\int_0^{\Delta t} \exp\{As\} \begin{pmatrix} 1 & 0 \\ 0 & 0.7071 \end{pmatrix} ds \right]}_B u[k], \quad x[0] = \begin{pmatrix} 0 \\ 0 \end{pmatrix}, \quad (3.7.12)$$

where $u[k]$ is the value of the control on the “continuous time” interval $(k\Delta t, (k+1)\Delta t)$.

We build the inner \mathcal{I}_k and the outer \mathcal{O}_k ellipsoidal approximations of the domains $D^k = D_{k\Delta t}$ in a recurrent manner:

- the ellipses \mathcal{I}_0 and \mathcal{O}_0 are just the singletons (the origin);
- \mathcal{I}_{k+1} is the best (the largest in the area) ellipse contained in the set

$$S\mathcal{I}_k + BW, \quad W = \{u \in \mathbf{R}^2 \mid \|u\|_2 \leq 1\},$$

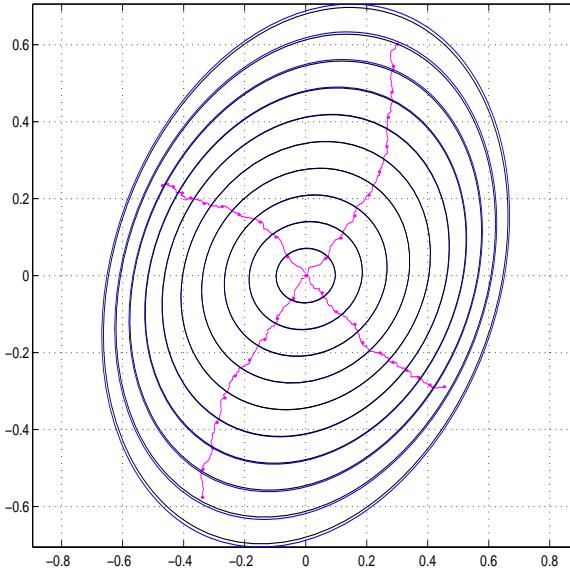
which is the sum of two ellipses;

- \mathcal{O}_{k+1} is the best (the smallest in the area) ellipse containing the set

$$S\mathcal{O}_k + BW,$$

which again is the sum of two ellipses.

Here is the picture we get:



Outer and inner approximations of the “reachability domains”

$$D^{10\ell} = D_{0.1\ell \text{ sec}}, \ell = 1, 2, \dots, 10, \text{ for system (3.7.12)}$$

- Ten pairs of ellipses are the outer and inner approximations of the domains D^1, \dots, D^{10} (look how close the ellipses from a pair are close to each other!);
- Four curves are sample trajectories of the system (dots correspond to time instants 0.1ℓ sec in continuous time, i.e., time instants 10ℓ in discrete time, $\ell = 0, 1, \dots, 10$).

3.8 Exercises

3.8.1 Around positive semidefiniteness, eigenvalues and \succeq -ordering

3.8.1.1 Criteria for positive semidefiniteness

Recall the criterion of positive definiteness of a symmetric matrix:

[Sylvester] A symmetric $m \times m$ matrix $A = [a_{ij}]_{i,j=1}^m$ is positive definite if and only if all angular minors

$$\text{Det} \left([a_{ij}]_{i,j=1}^k \right), \quad k = 1, \dots, m,$$

are positive.

Exercise 3.2 Prove that a symmetric $m \times m$ matrix A is positive semidefinite if and only if all its principal minors (i.e., determinants of square sub-matrices symmetric w.r.t. the diagonal) are nonnegative.

Hint: look at the angular minors of the matrices $A + \epsilon I_n$ for small positive ϵ .

Demonstrate by an example that nonnegativity of angular minors of a symmetric matrix is not sufficient for the positive semidefiniteness of the matrix.

Exercise 3.3 [Diagonal-dominant matrices] Let a symmetric matrix $A = [a_{ij}]_{i,j=1}^m$ satisfy the relation

$$a_{ii} \geq \sum_{j \neq i} |a_{ij}|, \quad i = 1, \dots, m.$$

Prove that A is positive semidefinite.

3.8.1.2 Variational characterization of eigenvalues

The basic fact about eigenvalues of a symmetric matrix is the following

Variational Characterization of Eigenvalues [Theorem A.7.3] Let A be a symmetric $m \times m$ matrix and $\lambda(A) = (\lambda_1(A), \dots, \lambda_m(A))$ be the vector of eigenvalues of A taken with their multiplicities and arranged in non-ascending order:

$$\lambda_1(A) \geq \lambda_2(A) \geq \dots \geq \lambda_m(A).$$

Then for every $i = 1, \dots, m$ one has:

$$\lambda_i(A) = \min_{E \in \mathcal{E}_i} \max_{v \in E, v^T v = 1} v^T A v,$$

where \mathcal{E}_i is the family of all linear subspaces of \mathbf{R}^m of dimension $m - i + 1$.

Singular values of rectangular matrices also admit variational description:

Variational Characterization of Singular Values Let A be an $m \times n$ matrix, $m \leq n$, and let $\sigma(A) = \lambda((AA^T)^{1/2})$ be the vector of singular values of A . Then for every $i = 1, \dots, m$ one has:

$$\sigma_i(A) = \min_{E \in \mathcal{E}_i} \max_{v \in E, v^T v = 1} \|Av\|_2,$$

where \mathcal{E}_i is the family of all linear subspaces of \mathbf{R}^n of dimension $n - i + 1$.

Exercise 3.4 Derive the Variational Characterization of Singular Values from the Variational Characterization of Eigenvalues.

Exercise 3.5 Derive from the Variational Characterization of Eigenvalues the following facts:

- (i) [Monotonicity of the vector of eigenvalues] If $A \succeq B$, then $\lambda(A) \geq \lambda(B)$;
- (ii) The functions $\lambda_1(X), \lambda_m(X)$ of $X \in \mathbf{S}^m$ are convex and concave, respectively.
- (iii) If Δ is a convex subset of the real axis, then the set of all matrices $X \in \mathbf{S}^m$ with spectrum from Δ is convex.

Recall now the definition of a function of symmetric matrix. Let A be a symmetric $m \times m$ matrix and

$$p(t) = \sum_{i=0}^k p_i t^i$$

be a real polynomial on the axis. By definition,

$$p(A) = \sum_{i=0}^k p_i A^i \in \mathbf{S}^m.$$

This definition is compatible with the arithmetic of real polynomials: when you add/multiply polynomials, you add/multiply the “values” of these polynomials at every fixed symmetric matrix:

$$(p + q)(A) = p(A) + q(A); (p \cdot q)(A) = p(A)q(A).$$

A nice feature of this definition is that

(A) For $A \in \mathbf{S}^m$, the matrix $p(A)$ depends only on the restriction of p on the spectrum (set of eigenvalues) of A : if p and q are two polynomials such that $p(\lambda_i(A)) = q(\lambda_i(A))$ for $i = 1, \dots, m$, then $p(A) = q(A)$.

Indeed, we can represent a symmetric matrix A as $A = U^T \Lambda U$, where U is orthogonal and Λ is diagonal with the eigenvalues of A on its diagonal. Since $UU^T = I$, we have $A^i = U^T \Lambda^i U$; consequently,

$$p(A) = U^T p(\Lambda) U,$$

and since the matrix $p(\Lambda)$ depends on the restriction of p on the spectrum of A only, the result follows.

As a byproduct of our reasoning, we get an “explicit” representation of $p(A)$ in terms of the spectral decomposition $A = U^T \Lambda U$ (U is orthogonal, Λ is diagonal with the diagonal $\lambda(A)$):

(B) The matrix $p(A)$ is just $U^T \text{Diag}(p(\lambda_1(A)), \dots, p(\lambda_n(A))) U$.

(A) allows to define arbitrary functions of matrices, not necessarily polynomials:

Let A be symmetric matrix and f be a real-valued function defined at least at the spectrum of A . By definition, the matrix $f(A)$ is defined as $p(A)$, where p is a polynomial coinciding with f on the spectrum of A . (The definition makes sense, since by **(A)** $p(A)$ depends only on the restriction of p on the spectrum of A , i.e., every “polynomial continuation” $p(\cdot)$ of f from the spectrum of A to the entire axis results in the same $p(A)$).

The “calculus of functions of a symmetric matrix” is fully compatible with the usual arithmetic of functions, e.g.:

$$(f + g)(A) = f(A) + g(A); (\mu f)(A) = \mu f(A); (f \cdot g)(A) = f(A)g(A); (f \circ g)(A) = f(g(A)),$$

provided that the functions in question are well-defined on the spectrum of the corresponding matrix. And of course the spectral decomposition of $f(A)$ is just $f(A) = U^T \text{Diag}(f(\lambda_1(A)), \dots, f(\lambda_m(A))) U$, where $A = U^T \text{Diag}(\lambda_1(A), \dots, \lambda_m(A)) U$ is the spectral decomposition of A .

Note that “Calculus of functions of symmetric matrices” becomes very unusual when we are trying to operate with functions of several (non-commuting) matrices. E.g., it is generally not true that $\exp\{A + B\} = \exp\{A\} \exp\{B\}$ (the right hand side matrix may be even non-symmetric!). It is also generally not true that if f is monotone and $A \succeq B$, then $f(A) \succeq f(B)$, etc.

Exercise 3.6 Demonstrate by an example that the relation $0 \preceq A \preceq B$ does not necessarily imply that $A^2 \preceq B^2$.

By the way, the relation $0 \preceq A \preceq B$ does imply that $0 \preceq A^{1/2} \preceq B^{1/2}$.

Sometimes, however, we can get “weak” matrix versions of usual arithmetic relations. E.g.,

Exercise 3.7 Let f be a nondecreasing function on the real line, and let $A \succeq B$. Prove that $\lambda(f(A)) \geq \lambda(f(B))$.

The strongest (and surprising) “weak” matrix version of a usual (“scalar”) inequality is as follows.

Let $f(t)$ be a *closed* convex function on the real line; by definition, it means that f is a function on the axis taking real values and the value $+\infty$ such that

- the set $\text{Dom } f$ of the values of argument where f is finite is convex and nonempty;

- if a sequence $\{t_i \in \text{Dom } f\}$ converges to a point t and the sequence $f(t_i)$ has a limit, then $t \in \text{Dom } f$ and $f(t) \leq \lim_{i \rightarrow \infty} f(t_i)$ (this property is called “lower semicontinuity”).

E.g., the function $f(x) = \begin{cases} 0, & 0 \leq t \leq 1 \\ +\infty, & \text{otherwise} \end{cases}$ is closed. In contrast to this, the functions

$$g(x) = \begin{cases} 0, & 0 < t \leq 1 \\ 1, & t = 0 \\ +\infty, & \text{for all remaining } t \end{cases}$$

and

$$h(x) = \begin{cases} 0, & 0 < t < 1 \\ +\infty, & \text{otherwise} \end{cases}$$

are not closed, although they are convex: a closed function cannot “jump up” at an endpoint of its domain, as it is the case for g , and it cannot take value $+\infty$ at a point, if it takes values $\leq a < \infty$ in a neighbourhood of the point, as it is the case for h .

For a convex function f , its *Legendre transformation* f_* (also called the *conjugate*, or the *Fenchel dual* of f) is defined as

$$f_*(s) = \sup_t [ts - f(t)].$$

It turns out that the Legendre transformation of a closed convex function also is closed and convex, and that twice taken Legendre transformation of a closed convex function is this function.

The Legendre transformation (which, by the way, can be defined for convex functions on \mathbf{R}^n as well) underlies many standard inequalities. Indeed, by definition of f_* we have

$$f_*(s) + f(t) \geq st \quad \forall s, t; \tag{L}$$

For specific choices of f , we can derive from the general inequality (L) many useful inequalities. E.g.,

- If $f(t) = \frac{1}{2}t^2$, then $f_*(s) = \frac{1}{2}s^2$, and (L) becomes the standard inequality

$$st \leq \frac{1}{2}t^2 + \frac{1}{2}s^2 \quad \forall s, t \in \mathbf{R};$$

- If $1 < p < \infty$ and $f(t) = \begin{cases} \frac{t^p}{p}, & t \geq 0 \\ +\infty, & t < 0 \end{cases}$, then $f_*(s) = \begin{cases} \frac{s^q}{q}, & s \geq 0 \\ +\infty, & s < 0 \end{cases}$, with q given by $\frac{1}{p} + \frac{1}{q} = 1$, and (L) becomes the Young inequality

$$\forall(s, t \geq 0) : ts \leq \frac{t^p}{p} + \frac{s^q}{q}, \quad 1 < p, q < \infty, \quad \frac{1}{p} + \frac{1}{q} = 1.$$

Now, what happens with (L) if s, t are symmetric matrices? Of course, both sides of (L) still make sense and are matrices, but we have no hope to say something reasonable about the relation between these matrices (e.g., the right hand side in (L) is not necessarily symmetric). However,

Exercise 3.8 *Let f_* be a closed convex function with the domain $\text{Dom } f_* \subset \mathbf{R}_+$, and let f be the Legendre transformation of f_* . Then for every pair of symmetric matrices X, Y of the same size with the spectrum of X belonging to $\text{Dom } f$ and the spectrum of Y belonging to $\text{Dom } f_*$ one has*

$$\lambda(f(X)) \geq \lambda(Y^{1/2}XY^{1/2} - f_*(Y))^{24)}$$

3.8.1.3 Birkhoff's Theorem

Surprisingly enough, one of the most useful facts about eigenvalues of symmetric matrices is the following, essentially combinatorial, statement (it does not mention the word “eigenvalue” at all).

Birkhoff's Theorem. Consider the set \mathcal{S}_m of double-stochastic $m \times m$ matrices, i.e., square matrices $[p_{ij}]_{i,j=1}^m$ satisfying the relations

$$\begin{aligned} p_{ij} &\geq 0, \quad i, j = 1, \dots, m; \\ \sum_{i=1}^m p_{ij} &= 1, \quad j = 1, \dots, m; \\ \sum_{j=1}^m p_{ij} &= 1, \quad i = 1, \dots, m. \end{aligned}$$

A matrix P belongs to \mathcal{S}_m if and only if it can be represented as a convex combination of $m \times m$ permutation matrices:

$$P \in \mathcal{S}_m \Leftrightarrow \exists (\lambda_i \geq 0, \sum_i \lambda_i = 1) : P = \sum_i \lambda_i \Pi^i,$$

where all Π^i are permutation matrices (i.e., with exactly one nonzero element, equal to 1, in every row and every column).

For proof, see Section B.2.8.C.1.

An immediate corollary of the Birkhoff Theorem is the following fact:

(C) Let $f : \mathbf{R}^m \rightarrow \mathbf{R} \cup \{+\infty\}$ be a convex symmetric function (symmetry means that the value of the function remains unchanged when we permute the coordinates in an argument), let $x \in \text{Dom } f$ and $P \in \mathcal{S}_m$. Then

$$f(Px) \leq f(x).$$

The proof is immediate: by Birkhoff's Theorem, Px is a convex combination of a number of permutations x^i of x . Since f is convex, we have

$$f(Px) \leq \max_i f(x^i) = f(x),$$

the concluding equality resulting from the symmetry of f .

²⁴⁾ In the scalar case, our inequality reads $f(x) \geq y^{1/2}xy^{1/2} - f_*(y)$, which is an equivalent form of (L) when $\text{Dom } f_* \subset \mathbf{R}_+$.

The role of **(C)** in numerous questions related to eigenvalues is based upon the following simple

Observation. Let A be a symmetric $m \times m$ matrix. Then the diagonal $\text{Dg}(A)$ of the matrix A is the image of the vector $\lambda(A)$ of the eigenvalues of A under multiplication by a double stochastic matrix:

$$\text{Dg}(A) = P\lambda(A) \text{ for some } P \in \mathcal{S}_m$$

Indeed, consider the spectral decomposition of A :

$$A = U^T \text{Diag}(\lambda_1(A), \dots, \lambda_m(A))U$$

with orthogonal $U = [u_{ij}]$. Then

$$A_{ii} = \sum_{j=1}^m u_{ji}^2 \lambda_j(A) \equiv (P\lambda(A))_i,$$

where the matrix $P = [u_{ji}^2]_{i,j=1}^m$ is double stochastic.

Combining the Observation and **(C)**, we conclude that if f is a convex symmetric function on \mathbf{R}^m , then for every $m \times m$ symmetric matrix A one has

$$f(\text{Dg}(A)) \leq f(\lambda(A)).$$

Moreover, let \mathcal{O}_m be the set of all orthogonal $m \times m$ matrices. For every $V \in \mathcal{O}_m$, the matrix $V^T A V$ has the same eigenvalues as A , so that for a convex symmetric f one has

$$f(\text{Dg}(V^T A V)) \leq f(\lambda(V^T A V)) = f(\lambda(A)),$$

whence

$$f(\lambda(A)) \geq \max_{V \in \mathcal{O}_m} f(\text{Dg}(V^T A V)).$$

In fact the inequality here is equality, since for properly chosen $V \in \mathcal{O}_m$ we have $\text{Dg}(V^T A V) = \lambda(A)$. We have arrived at the following result:

(D) Let f be a symmetric convex function on \mathbf{R}^m . Then for every symmetric $m \times m$ matrix A one has

$$f(\lambda(A)) = \max_{V \in \mathcal{O}_m} f(\text{Dg}(V^T A V)),$$

\mathcal{O}_m being the set of all $m \times m$ orthogonal matrices.

In particular, the function

$$F(A) = f(\lambda(A))$$

is convex in $A \in \mathbf{S}^m$ (as the maximum of a family of convex in A functions $F_V(A) = f(\text{Dg}(V^T A V))$, $V \in \mathcal{O}_m$.)

Exercise 3.9 Let $g(t) : \mathbf{R} \rightarrow \mathbf{R} \cup \{+\infty\}$ be a convex function, and let \mathcal{F}_n be the set of all matrices $X \in \mathbf{S}^n$ with the spectrum belonging to $\text{Dom } g$. Prove that the function $\text{Tr}(g(X))$ is convex on \mathcal{F}_n .

Hint: Apply **(D)** to the function $f(x_1, \dots, x_n) = g(x_1) + \dots + g(x_n)$.

Exercise 3.10 Let $A = [a_{ij}]$ be a symmetric $m \times m$ matrix. Prove that

$$(i) \text{ Whenever } p \geq 1, \text{ one has } \sum_{i=1}^m |a_{ii}|^p \leq \sum_{i=1}^m |\lambda_i(A)|^p;$$

$$(ii) \text{ Whenever } A \text{ is positive semidefinite, } \prod_{i=1}^m a_{ii} \geq \text{Det}(A);$$

(iii) For $x \in \mathbf{R}^m$, let the function $S_k(x)$ be the sum of k largest entries of x (i.e., the sum of the first k entries in the vector obtained from x by writing down the coordinates of x in the non-ascending order). Prove that $S_k(x)$ is a convex symmetric function of x and derive from this observation that

$$S_k(\text{Dg}(A)) \leq S_k(\lambda(A)).$$

Hint: note that $S_k(x) = \max_{1 \leq i_1 < i_2 < \dots < i_k \leq m} \sum_{l=1}^k x_{i_l}$.

(iv) [Trace inequality] Whenever $A, B \in \mathbf{S}^m$, one has

$$\lambda^T(A)\lambda(B) \geq \text{Tr}(AB).$$

Exercise 3.11 Prove that if $A \in \mathbf{S}^m$ and $p, q \in [1, \infty]$ are such that $\frac{1}{p} + \frac{1}{q} = 1$, then

$$\max_{B \in \mathbf{S}^m : \|\lambda(B)\|_q = 1} \text{Tr}(AB) = \|\lambda(A)\|_p.$$

In particular, $\|\lambda(\cdot)\|_p$ is a norm on \mathbf{S}^m , and the conjugate of this norm is $\|\lambda(\cdot)\|_q$, $\frac{1}{p} + \frac{1}{q} = 1$.

Exercise 3.12 Let $X = \begin{pmatrix} X_{11} & X_{12} & \dots & X_{1m} \\ X_{12}^T & X_{22} & \dots & X_{2m} \\ \vdots & \vdots & \ddots & \dots \\ X_{1m}^T & X_{2m}^T & \dots & X_{mm} \end{pmatrix}$ be an $n \times n$ symmetric matrix which is

partitioned into m^2 blocks X_{ij} in a symmetric, w.r.t. the diagonal, fashion (so that the blocks X_{jj} are square), and let

$$\hat{X} = \begin{pmatrix} X_{11} & & & \\ & X_{22} & & \\ & & \ddots & \\ & & & X_{mm} \end{pmatrix}.$$

1) Let $F : \mathbf{S}^n \rightarrow \mathbf{R} \cup \{+\infty\}$ be a convex “rotation-invariant” function: for all $Y \in \mathbf{S}^n$ and all orthogonal matrices U one has $F(U^T Y U) = F(Y)$. Prove that

$$F(\hat{X}) \leq F(X).$$

Hint: Represent the matrix \hat{X} as a convex combination of the rotations $U^T X U$, $U^T U = I$, of X .

2) Let $f : \mathbf{R}^n \rightarrow \mathbf{R} \cup \{+\infty\}$ be a convex symmetric w.r.t. permutations of the entries in the argument function, and let $F(Y) = f(\lambda(Y))$, $Y \in \mathbf{S}^n$. Prove that

$$F(\hat{X}) \leq F(X).$$

3) Let $g : \mathbf{R} \rightarrow \mathbf{R} \cup \{+\infty\}$ be convex function on the real line which is finite on the set of eigenvalues of X , and let $\mathcal{F}_n \subset \mathbf{S}^n$ be the set of all $n \times n$ symmetric matrices with all eigenvalues belonging to the domain of g . Assume that the mapping

$$Y \mapsto g(Y) : \mathcal{F}_n \rightarrow \mathbf{S}^n$$

is \succeq -convex:

$$g(\lambda' Y' + \lambda'' Y'') \preceq \lambda' g(Y') + \lambda'' g(Y'') \quad \forall (Y', Y'') \in \mathcal{F}_n, \lambda', \lambda'' \geq 0, \lambda' + \lambda'' = 1.$$

Prove that

$$(g(X))_{ii} \succeq g(X_{ii}), \quad i = 1, \dots, m,$$

where the partition of $g(X)$ into the blocks $(g(X))_{ij}$ is identical to the partition of X into the blocks X_{ij} .

Exercise 3.12 gives rise to a number of interesting inequalities. Let X, \hat{X} be the same as in the Exercise, and let $[Y]$ denote the northwest block, of the same size as X_{11} , of an $n \times n$ matrix Y . Then

$$1. \left(\sum_{i=1}^m \|\lambda(X_{ii})\|_p^p \right)^{1/p} \leq \|\lambda(X)\|_p, \quad 1 \leq p < \infty$$

[Exercise 3.12.2), $f(x) = \|x\|_p$];

$$2. \text{ If } X \succ 0, \text{ then } \text{Det}(X) \leq \prod_{i=1}^m \text{Det}(X_{ii})$$

[Exercise 3.12.2), $f(x) = -(x_1 \dots x_n)^{1/n}$ for $x \geq 0$];

$$3. [X^2] \succeq X_{11}^2$$

[This inequality is nearly evident; it follows also from Exercise 3.12.3) with $g(t) = t^2$ (the \succeq -convexity of $g(Y)$ is stated in Exercise 3.22.1))];

$$4. \text{ If } X \succ 0, \text{ then } X_{11}^{-1} \preceq [X^{-1}]$$

[Exercise 3.12.3) with $g(t) = t^{-1}$ for $t > 0$; the \succeq -convexity of $g(Y)$ on \mathbf{S}_{++}^n is stated by Exercise 3.22.2)];

$$5. \text{ For every } X \succeq 0, [X^{1/2}] \preceq X_{11}^{1/2}$$

[Exercise 3.12.3) with $g(t) = -\sqrt{t}$; the \succeq -convexity of $g(Y)$ is stated by Exercise 3.22.4)].

Extension: If $X \succeq 0$, then for every $\alpha \in (0, 1)$ one has $[X^\alpha] \preceq X_{11}^\alpha$

[Exercise 3.12.3) with $g(t) = -t^\alpha$; the function $-Y^\alpha$ of $Y \succeq 0$ is known to be \succeq -convex];

$$6. \text{ If } X \succ 0, \text{ then } [\ln(X)] \preceq \ln(X_{11})$$

[Exercise 3.12.3) with $g(t) = -\ln t$, $t > 0$; the \succeq -convexity of $g(Y)$ is stated by Exercise 3.22.5)].

Exercise 3.13 1) Let $A = [a_{ij}]_{i,j} \succeq 0$, let $\alpha \geq 0$, and let $B \equiv [b_{ij}]_{i,j} = A^\alpha$. Prove that

$$b_{ii} \begin{cases} \leq a_{ii}^\alpha, & \alpha \leq 1 \\ \geq a_{ii}^\alpha, & \alpha \geq 1 \end{cases}$$

- 2) Let $A = [a_{ij}]_{i,j} \succ 0$, and let $B \equiv [b_{ij}]_{i,j} = A^{-1}$. Prove that $b_{ii} \geq a_{ii}^{-1}$.
 3) Let $[A]$ denote the northwest 2×2 block of a square matrix. Which of the implications

$$(a) \quad A \succeq 0 \Rightarrow [A^4] \succeq [A]^4$$

$$(b) \quad A \succeq 0 \Rightarrow [A^4]^{1/4} \succeq [A]$$

are true?

3.8.1.4 Semidefinite representations of functions of eigenvalues

The goal of the subsequent series of exercises is to prove Proposition 3.2.1.

We start with a description (important by its own right) of the convex hull of permutations of a given vector. Let $x \in \mathbf{R}^m$, and let $X[x]$ be the set of all convex combinations of $m!$ vectors obtained from x by all permutations of the coordinates.

Claim: [“Majorization principle”] $X[x]$ is exactly the solution set of the following system of inequalities in variables $y \in \mathbf{R}^m$:

$$\begin{aligned} S_j(y) &\leq S_j(x), \quad j = 1, \dots, m-1 \\ y_1 + \dots + y_m &= x_1 + \dots + x_m \end{aligned} \tag{+}$$

(recall that $S_j(y)$ is the sum of the largest j entries of a vector y).

Exercise 3.14 [Easy part of the claim] Let Y be the solution set of (+). Prove that $Y \supset X[x]$.

Hint: Use (C) and the convexity of the functions $S_j(\cdot)$.

Exercise 3.15 [Difficult part of the claim] Let Y be the solution set of (+). Prove that $Y \subset X[x]$.

Sketch of the proof: Let $y \in Y$. We should prove that $y \in X[x]$. By symmetry, we may assume that the vectors x and y are ordered: $x_1 \geq x_2 \geq \dots \geq x_m$, $y_1 \geq y_2 \geq \dots \geq y_m$. Assume that $y \notin X[x]$, and let us lead this assumption to a contradiction.

- 1) Since $X[x]$ clearly is a convex compact set and $y \notin X[x]$, there exists a linear functional $c(z) = \sum_{i=1}^m c_i z_i$ which separates y and $X[x]$:

$$c(y) > \max_{z \in X[x]} c(z).$$

Prove that such a functional can be chosen “to be ordered”: $c_1 \geq c_2 \geq \dots \geq c_m$.

- 2) Verify that

$$c(y) \equiv \sum_{i=1}^m c_i y_i = \sum_{i=1}^{m-1} (c_i - c_{i+1}) \sum_{j=1}^i y_j + c_m \sum_{j=1}^m y_j$$

(Abel’s formula – a discrete version of integration by parts). Use this observation along with “orderedness” of $c(\cdot)$ and the inclusion $y \in Y$ to conclude that $c(y) \leq c(x)$, thus coming to the desired contradiction.

Exercise 3.16 Use the Majorization principle to prove Proposition 3.2.1.

The next pair of exercises is aimed at proving Proposition 3.2.2.

Exercise 3.17 Let $x \in \mathbf{R}^m$, and let $X_+[x]$ be the set of all vectors x' dominated by a vector from $X[x]$:

$$X_+[x] = \{y \mid \exists z \in X[x] : y \leq z\}.$$

- 1) Prove that $X_+[x]$ is a closed convex set.
- 2) Prove the following characterization of $X_+[x]$:

$X_+[x]$ is exactly the set of solutions of the system of inequalities $S_j(y) \leq S_j(x)$, $j = 1, \dots, m$, in variables y .

Hint: Modify appropriately the construction outlined in Exercise 3.15.

Exercise 3.18 Derive Proposition 3.2.2 from the result of Exercise 3.17.2).

3.8.1.5 Cauchy's inequality for matrices

The standard Cauchy's inequality says that

$$\left| \sum_i x_i y_i \right| \leq \sqrt{\sum_i x_i^2} \sqrt{\sum_i y_i^2} \quad (3.8.1)$$

for reals x_i, y_i , $i = 1, \dots, n$; this inequality is exact in the sense that for every collection x_1, \dots, x_n there exists a collection y_1, \dots, y_n with $\sum_i y_i^2 = 1$ which makes (3.8.1) an equality.

Exercise 3.19 (i) Prove that whenever $X_i, Y_i \in \mathbf{M}^{p,q}$, one has

$$\sigma(\sum_i X_i^T Y_i) \leq \lambda \left(\left[\sum_i X_i^T X_i \right]^{1/2} \right) \|\lambda(\sum_i Y_i^T Y_i)\|_\infty^{1/2} \quad (*)$$

where $\sigma(A) = \lambda([AA^T]^{1/2})$ is the vector of singular values of a matrix A arranged in the non-ascending order.

Prove that for every collection $X_1, \dots, X_n \in \mathbf{M}^{p,q}$ there exists a collection $Y_1, \dots, Y_n \in \mathbf{M}^{p,q}$ with $\sum_i Y_i^T Y_i = I_q$ which makes (*) an equality.

(ii) Prove the following “matrix version” of the Cauchy inequality: whenever $X_i, Y_i \in \mathbf{M}^{p,q}$, one has

$$\left| \sum_i \text{Tr}(X_i^T Y_i) \right| \leq \text{Tr} \left(\left[\sum_i X_i^T X_i \right]^{1/2} \right) \|\lambda(\sum_i Y_i^T Y_i)\|_\infty^{1/2}, \quad (**)$$

and for every collection $X_1, \dots, X_n \in \mathbf{M}^{p,q}$ there exists a collection $Y_1, \dots, Y_n \in \mathbf{M}^{p,q}$ with $\sum_i Y_i^T Y_i = I_q$ which makes (**) an equality.

Here is another exercise of the same flavour:

Exercise 3.20 For nonnegative reals a_1, \dots, a_m and a real $\alpha > 1$ one has

$$\left(\sum_{i=1}^m a_i^\alpha \right)^{1/\alpha} \leq \sum_{i=1}^m a_i.$$

Both sides of this inequality make sense when the nonnegative reals a_i are replaced with positive semidefinite $n \times n$ matrices A_i . What happens with the inequality in this case?

Consider the following four statements (where $\alpha > 1$ is a real and $m, n > 1$):

1)

$$\forall(A_i \in \mathbf{S}_+^n) : \left(\sum_{i=1}^m A_i^\alpha \right)^{1/\alpha} \preceq \sum_{i=1}^m A_i.$$

2)

$$\forall(A_i \in \mathbf{S}_+^n) : \lambda_{\max} \left(\left(\sum_{i=1}^m A_i^\alpha \right)^{1/\alpha} \right) \leq \lambda_{\max} \left(\sum_{i=1}^m A_i \right).$$

3)

$$\forall(A_i \in \mathbf{S}_+^n) : \text{Tr} \left(\left(\sum_{i=1}^m A_i^\alpha \right)^{1/\alpha} \right) \leq \text{Tr} \left(\sum_{i=1}^m A_i \right).$$

4)

$$\forall(A_i \in \mathbf{S}_+^n) : \text{Det} \left(\left(\sum_{i=1}^m A_i^\alpha \right)^{1/\alpha} \right) \leq \text{Det} \left(\sum_{i=1}^m A_i \right).$$

Among these 4 statements, exactly 2 are true. Identify and prove the true statements.

3.8.1.6 \succeq -convexity of some matrix-valued functions

Consider a function $F(x)$ defined on a convex set $X \subset \mathbf{R}^n$ and taking values in \mathbf{S}^m . We say that such a function is \succeq -convex, if

$$F(\alpha x + (1 - \alpha)y) \preceq \alpha F(x) + (1 - \alpha)F(y)$$

for all $x, y \in X$ and all $\alpha \in [0, 1]$. F is called \succeq -concave, if $-F$ is \succeq -convex.

A function $F : \text{Dom } F \rightarrow \mathbf{S}^m$ defined on a set $\text{Dom } F \subset \mathbf{S}^k$ is called \succeq -monotone, if

$$x, y \in \text{Dom } F, x \succeq y \Rightarrow F(x) \succeq F(y);$$

F is called \succeq -antimonotone, if $-F$ is \succeq -monotone.

Exercise 3.21 1) Prove that a function $F : X \rightarrow \mathbf{S}^m$, $X \subset \mathbf{R}^n$, is \succeq -convex if and only if its “epigraph”

$$\{(x, Y) \in \mathbf{R}^n \rightarrow \mathbf{S}^m \mid x \in X, F(x) \preceq Y\}$$

is a convex set.

2) Prove that a function $F : X \rightarrow \mathbf{S}^m$ with convex domain $X \subset \mathbf{R}^n$ is \succeq -convex if and only if for every $A \in \mathbf{S}_+^m$ the function $\text{Tr}(AF(x))$ is convex on X .

3) Let $X \subset \mathbf{R}^n$ be a convex set with a nonempty interior and $F : X \rightarrow \mathbf{S}^m$ be a function continuous on X which is twice differentiable in $\text{int } X$. Prove that F is \succeq -convex if and only if the second directional derivative of F

$$D^2F(x)[h, h] \equiv \frac{d^2}{dt^2} \Big|_{t=0} F(x + th)$$

is $\succeq 0$ for every $x \in \text{int } X$ and every direction $h \in \mathbf{R}^n$.

4) Let $F : \text{dom}F \rightarrow \mathbf{S}^m$ be defined and continuously differentiable on an open convex subset of \mathbf{S}^k . Prove that the necessary and sufficient condition for F to be \succeq -monotone is the validity of the implication

$$h \in \mathbf{S}_+^k, x \in \text{Dom } F \Rightarrow DF(x)[h] \succeq 0.$$

5) Let F be \succeq -convex and $\mathcal{S} \subset \mathbf{S}^m$ be a convex set which is \succeq -antimonotone, i.e. whenever $Y' \preceq Y$ and $Y \in \mathcal{S}$, one has $Y' \in \mathcal{S}$. Prove that the set $F^{-1}(\mathcal{S}) = \{x \in X \mid F(x) \in \mathcal{S}\}$ is convex.

6) Let $G : \text{Dom } G \rightarrow \mathbf{S}^k$ and $F : \text{Dom } F \rightarrow \mathbf{S}^m$, let $G(\text{Dom } G) \subset \text{Dom } F$, and let $H(x) = F(G(x)) : \text{Dom } G \rightarrow \mathbf{S}^m$.

a) Prove that if G and F are \succeq -convex and F is \succeq -monotone, then H is \succeq -convex.

b) Prove that if G and F are \succeq -concave and F is \succeq -monotone, then H is \succeq -concave.

7) Let $F_i : G \rightarrow \mathbf{S}^m$, and assume that for every $x \in G$ exists

$$F(x) = \lim_{i \rightarrow \infty} F_i(x).$$

Prove that if all functions from the sequence $\{F_i\}$ are (a) \succeq -convex, or (b) \succeq -concave, or (c) \succeq -monotone, or (d) \succeq -antimonotone, then so is F .

The goal of the next exercise is to establish the \succeq -convexity of several matrix-valued functions.

Exercise 3.22 Prove that the following functions are \succeq -convex:

1) $F(x) = xx^T : \mathbf{M}^{p,q} \rightarrow \mathbf{S}^p$;

2) $F(x) = x^{-1} : \text{int } \mathbf{S}_+^m \rightarrow \text{int } \mathbf{S}_+^m$;

3) $F(u, v) = u^T v^{-1} u : \mathbf{M}^{p,q} \times \text{int } \mathbf{S}_+^p \rightarrow \mathbf{S}^q$;

Prove that the following functions are \succeq -concave and monotone:

4) $F(x) = x^{1/2} : \mathbf{S}_+^m \rightarrow \mathbf{S}^m$;

5) $F(x) = \ln x : \text{int } \mathbf{S}_+^m \rightarrow \mathbf{S}^m$;

6) $F(x) = (Ax^{-1}A^T)^{-1} : \text{int } \mathbf{S}_+^n \rightarrow \mathbf{S}^m$, provided that A is an $m \times n$ matrix of rank m .

3.8.2 SD representations of epigraphs of convex polynomials

Mathematically speaking, the central question concerning the “expressive abilities” of Semidefinite Programming is how wide is the family of convex sets which are SDr. By definition, an SDr set is the projection of the inverse image of \mathbf{S}_+^m under affine mapping. In other words, every SDr set is a projection of a convex set given by a number of polynomial inequalities (indeed, the cone \mathbf{S}_+^m is a convex set given by polynomial inequalities saying that all principal minors of matrix are nonnegative). Consequently, the inverse image of \mathbf{S}_+^m under an affine mapping is also a convex set given by a number of (non-strict) polynomial inequalities. And it is known that every projection of such a set is also given by a number of polynomial inequalities (both strict and non-strict). We conclude that

A SD-representable set always is a convex set given by finitely many polynomial inequalities (strict and non-strict).

A natural (and seemingly very difficult) question is whether the inverse is true – whether a convex set given by a number of polynomial inequalities is always SDr. This question can be simplified in many ways – we may fix the dimension of the set, we may assume the polynomials participating in inequalities to be convex, we may fix the degrees of the polynomials, etc.; to the best of our knowledge, all these questions are open.

The goal of the subsequent exercises is to answer affirmatively the simplest question of the above series:

Let $\pi(x)$ be a convex polynomial of one variable. Then its epigraph

$$\{(t, x) \in \mathbf{R}^2 \mid t \geq \pi(x)\}$$

is SDr.

Let us fix a nonnegative integer k and consider the curve

$$p(x) = (1, x, x^2, \dots, x^{2k})^T \in \mathbf{R}^{2k+1}.$$

Let Π_k be the closure of the convex hull of values of the curve. How can one describe Π_k ?

A convenient way to answer this question is to pass to a *matrix representation* of all objects involved. Namely, let us associate with a vector $\xi = (\xi_0, \xi_1, \dots, \xi_{2k}) \in \mathbf{R}^{2k+1}$ the $(k+1) \times (k+1)$ symmetric matrix

$$\mathcal{M}(\xi) = \begin{pmatrix} \xi_0 & \xi_1 & \xi_2 & \xi_3 & \cdots & \xi_k \\ \xi_1 & \xi_2 & \xi_3 & \xi_4 & \cdots & \xi_{k+1} \\ \xi_2 & \xi_3 & \xi_4 & \xi_5 & \cdots & \xi_{k+2} \\ \xi_3 & \xi_4 & \xi_5 & \xi_6 & \cdots & \xi_{k+3} \\ \vdots & \vdots & \vdots & \vdots & \ddots & \vdots \\ \xi_k & \xi_{k+1} & \xi_{k+2} & \xi_{k+3} & \cdots & \xi_{2k} \end{pmatrix},$$

so that

$$[\mathcal{M}(\xi)]_{ij} = \xi_{i+j}, \quad i, j = 0, \dots, k.$$

The transformation $\xi \mapsto \mathcal{M}(\xi) : \mathbf{R}^{2k+1} \rightarrow \mathbf{S}^{k+1}$ is a linear embedding; the image of Π_k under this embedding is the closure of the convex hull of values of the curve

$$P(x) = \mathcal{M}(p(x)).$$

It follows that the image $\widehat{\Pi}_k$ of Π_k under the mapping \mathcal{M} possesses the following properties:

(i) $\widehat{\Pi}_k$ belongs to the image of \mathcal{M} , i.e., to the subspace H_k of \mathbf{S}^{2k+1} comprised of *Hankel* matrices – matrices with entries depending on the sum of indices only:

$$H_k = \left\{ X \in \mathbf{S}^{2k+1} \mid i + j = i' + j' \Rightarrow X_{ij} = X_{i'j'} \right\};$$

(ii) $\widehat{\Pi}_k \subset \mathbf{S}_+^{k+1}$ (indeed, every matrix $\mathcal{M}(p(x))$ is positive semidefinite);

(iii) For every $X \in \widehat{\Pi}_k$ one has $X_{00} = 1$.

It turns out that properties (i) – (iii) characterize $\widehat{\Pi}_k$:

(G) A symmetric $(k+1) \times (k+1)$ matrix X belongs to $\widehat{\Pi}_k$ if and only if it possesses the properties (i) – (iii): its entries depend on sum of indices only (i.e., $X \in H_k$), X is positive semidefinite and $X_{00} = 1$.

(G) is a particular case of the classical results related to the so called “moment problem”. The goal of the subsequent exercises is to give a simple alternative proof of this statement.

Note that the mapping $\mathcal{M}^* : \mathbf{S}^{k+1} \rightarrow \mathbf{R}^{2k+1}$ conjugate to the mapping \mathcal{M} is as follows:

$$(\mathcal{M}^* X)_l = \sum_{i=0}^l X_{i,l-i}, \quad l = 0, 1, \dots, 2k,$$

and we know something about this mapping: Example 21a (Lecture 3) says that

(H) The image of the cone \mathbf{S}_+^{k+1} under the mapping \mathcal{M}^* is exactly the cone of coefficients of polynomials of degree $\leq 2k$ which are nonnegative on the entire real line.

Exercise 3.23 Derive **(G)** from **(H)**.

(G), among other useful things, implies the result we need:

(I) Let $\pi(x) = \pi_0 + \pi_1x + \pi_2x^2 + \dots + \pi_{2k}x^{2k}$ be a convex polynomial of degree $2k$. Then the epigraph of π is SDr:

$$\{(t, x) \in \mathbf{R}^2 \mid t \geq p(x)\} = \mathcal{X}[\pi],$$

where

$$\mathcal{X}[\pi] = \left\{ (t, x) \mid \begin{array}{l} \exists x_2, \dots, x_{2k} : \begin{pmatrix} 1 & x & x_2 & x_3 & \dots & x_k \\ x & x_2 & x_3 & x_4 & \dots & x_{k+1} \\ x_2 & x_3 & x_4 & x_5 & \dots & x_{k+2} \\ x_3 & x_4 & x_5 & x_6 & \dots & x_{k+3} \\ \dots & \dots & \dots & \dots & \ddots & \dots \\ x_k & x_{k+1} & x_{k+2} & x_{k+3} & \dots & x_{2k} \end{pmatrix} \succeq 0, \\ \pi_0 + \pi_1x + \pi_2x_2 + \pi_3x_3 + \dots + \pi_{2k}x_{2k} \leq t \end{array} \right\}$$

Exercise 3.24 Prove **(I)**.

Note that the set $\mathcal{X}[\pi]$ makes sense for an arbitrary polynomial π , not necessary for a convex one. What is the projection of this set onto the (t, x) -plane? The answer is surprisingly nice: this is the convex hull of the epigraph of the polynomial π !

Exercise 3.25 Let $\pi(x) = \pi_0 + \pi_1x + \dots + \pi_{2k}x^{2k}$ with $\pi_{2k} > 0$, and let

$$G[\pi] = \text{Conv}\{(t, x) \in \mathbf{R}^2 \mid t \geq p(x)\}$$

be the convex hull of the epigraph of π (the set of all convex combinations of points from the epigraph of π).

- 1) Prove that $G[\pi]$ is a closed convex set.
- 2) Prove that

$$G[\pi] = \mathcal{X}[\pi].$$

3.8.3 Around the Lovasz capacity number and semidefinite relaxations of combinatorial problems

Recall that the Lovasz capacity number $\Theta(\Gamma)$ of an n -node graph Γ is the optimal value of the following semidefinite program:

$$\min_{\lambda, x} \{\lambda : \lambda I_n - \mathcal{L}(x) \succeq 0\} \tag{L}$$

where the symmetric $n \times n$ matrix $\mathcal{L}(x)$ is defined as follows:

- the dimension of x is equal to the number of arcs in Γ , and the coordinates of x are indexed by these arcs;

- the element of $\mathcal{L}(x)$ in an “empty” cell ij (one for which the nodes i and j are not linked by an arc in Γ) is 1;
- the elements of $\mathcal{L}(x)$ in a pair of symmetric “non-empty” cells ij, ji (those for which the nodes i and j are linked by an arc) are equal to the coordinate of x indexed by the corresponding arc.

As we remember, the importance of $\Theta(\Gamma)$ comes from the fact that $\Theta(\Gamma)$ is a computable upper bound on the stability number $\alpha(\Gamma)$ of the graph. We have seen also that the Shor semidefinite relaxation of the problem of finding the stability number of Γ leads to a “seemingly stronger” upper bound on $\alpha(\Gamma)$, namely, the optimal value $\sigma(\Gamma)$ in the semidefinite program

$$\min_{\lambda, \mu, \nu} \left\{ \lambda : \begin{pmatrix} \lambda & -\frac{1}{2}(e + \mu)^T \\ -\frac{1}{2}(e + \mu) & A(\mu, \nu) \end{pmatrix} \succeq 0 \right\} \quad (\text{Sh})$$

where $e = (1, \dots, 1)^T \in \mathbf{R}^n$ and $A(\mu, \nu)$ is the matrix as follows:

- the dimension of ν is equal to the number of arcs in Γ , and the coordinates of ν are indexed by these arcs;
- the diagonal entries of $A(\mu, \nu)$ are μ_1, \dots, μ_n ;
- the off-diagonal entries of $A(\mu, \nu)$ corresponding to “empty cells” are zeros;
- the off-diagonal entries of $A(\mu, \nu)$ in a pair of symmetric “non-empty” cells ij, ji are equal to the coordinate of ν indexed by the corresponding arc.

We have seen that (L) can be obtained from (Sh) when the variables μ_i are set to 1, so that $\sigma(\Gamma) \leq \Theta(\Gamma)$. Thus,

$$\alpha(\Gamma) \leq \sigma(\Gamma) \leq \Theta(\Gamma). \quad (3.8.2)$$

Exercise 3.26 1) Prove that if (λ, μ, ν) is a feasible solution to (Sh), then there exists a symmetric $n \times n$ matrix A such that $\lambda I_n - A \succeq 0$ and at the same time the diagonal entries of A and the off-diagonal entries in the “empty cells” are ≥ 1 . Derive from this observation that the optimal value in (Sh) is not less than the optimal value $\Theta'(\Gamma)$ in the following semidefinite program:

$$\min_{\lambda, x} \{ \lambda : \lambda I_n - X \succeq 0, X_{ij} \geq 1 \text{ whenever } i, j \text{ are not adjacent in } \Gamma \} \quad (\text{Sc}).$$

- 2) Prove that $\Theta'(\Gamma) \geq \alpha(\Gamma)$.

Hint: Demonstrate that if all entries of a symmetric $k \times k$ matrix are ≥ 1 , then the maximum eigenvalue of the matrix is at least k . Derive from this observation and the Interlacing Eigenvalues Theorem (Exercise 3.5.(ii)) that if a symmetric matrix contains a principal $k \times k$ submatrix with entries ≥ 1 , then the maximum eigenvalue of the matrix is at least k .

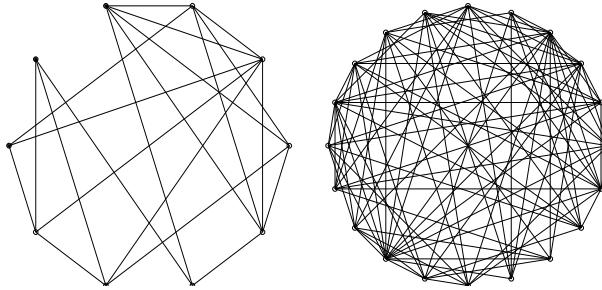
The upper bound $\Theta'(\Gamma)$ on the stability number of Γ is called the *Schrijver capacity* of graph Γ . Note that we have

$$\alpha(\Gamma) \leq \Theta'(\Gamma) \leq \sigma(\Gamma) \leq \Theta(\Gamma).$$

A natural question is which inequalities in this chain may happen to be strict. In order to answer it, we have computed the quantities in question for about 2,000 random graphs with number of

nodes varying 8 to 20. In our experiments, the stability number was computed – by brute force – for graphs with ≤ 12 nodes; for all these graphs, the integral part of $\Theta(\Gamma)$ was equal to $\alpha(\Gamma)$. Furthermore, $\Theta(\Gamma)$ was non-integer in 156 of our 2,000 experiments, and in 27 of these 156 cases the Schrijver capacity number $\Theta'(\Gamma)$ was strictly less than $\Theta(\Gamma)$. The quantities $\Theta'(\cdot), \sigma(\cdot), \Theta(\cdot)$ for 13 of these 27 cases are listed in the table below:

Graph #	# of nodes	α	Θ'	σ	Θ
1	20	?	4.373	4.378	4.378
2	20	?	5.062	5.068	5.068
3	20	?	4.383	4.389	4.389
4	20	?	4.216	4.224	4.224
5	13	?	4.105	4.114	4.114
6	20	?	5.302	5.312	5.312
7	20	?	6.105	6.115	6.115
8	20	?	5.265	5.280	5.280
9	9	3	3.064	3.094	3.094
10	12	4	4.197	4.236	4.236
11	8	3	3.236	3.302	3.302
12	12	4	4.236	4.338	4.338
13	10	3	3.236	3.338	3.338



Graphs # 13 (left) and # 8 (right); all nodes are on circumferences.

Exercise 3.27 Compute the stability numbers of the graphs # 8 and # 13.

Exercise 3.28 Prove that $\sigma(\Gamma) = \Theta(\Gamma)$.

The *chromatic number* $\xi(\Gamma)$ of a graph Γ is the minimal number of colours such that one can colour the nodes of the graph in such a way that no two adjacent (i.e., linked by an arc) nodes get the same colour²⁵⁾. The *complement* $\bar{\Gamma}$ of a graph Γ is the graph with the same set of nodes, and two distinct nodes in $\bar{\Gamma}$ are linked by an arc if and only if they are *not* linked by an arc in Γ .

²⁵⁾ E.g., when colouring a geographic map, it is convenient not to use the same colour for a pair of countries with a common border. It was observed that to meet this requirement for actual maps, 4 colours are sufficient. The famous “4-colour” Conjecture claims that this is so for every geographic map. Mathematically, you can represent a map by a graph, where the nodes represent the countries, and two nodes are linked by an arc if and only if the corresponding countries have common border. A characteristic feature of such a graph is that it is *planar* – you may draw it on 2D plane in such a way that the arcs will not cross each other, meeting only at the nodes. Thus, mathematical form of the 4-colour Conjecture is that the chromatic number of any planar graph is at most 4. This is indeed true, but it took about 100 years to prove the conjecture!

Lovasz proved that for every graph

$$\Theta(\Gamma) \leq \xi(\bar{\Gamma}) \quad (*)$$

so that

$$\alpha(\Gamma) \leq \Theta(\Gamma) \leq \xi(\bar{\Gamma})$$

(Lovasz's Sandwich Theorem).

Exercise 3.29 *Prove (*)*.

Hint: Let us colour the vertices of Γ in $k = \xi(\bar{\Gamma})$ colours in such a way that no two vertices of the same colour are adjacent in $\bar{\Gamma}$, i.e., every two nodes of the same colour are adjacent in Γ . Set $\lambda = k$, and let x be such that

$$[\mathcal{L}(x)]_{ij} = \begin{cases} -(k-1), & i \neq j, i, j \text{ are of the same colour} \\ 1, & \text{otherwise} \end{cases}$$

Prove that (λ, x) is a feasible solution to (L).

Now let us switch from the Lovasz capacity number to semidefinite relaxations of combinatorial problems, specifically to those of maximizing a quadratic form over the vertices of the unit cube, and over the entire cube:

$$\begin{aligned} (a) \quad & \max_x \left\{ x^T A x : x \in \text{Vrt}(C_n) = \{x \in \mathbf{R}^n \mid x_i = \pm 1 \forall i\} \right\} \\ (b) \quad & \max_x \left\{ x^T A x : x \in C_n = \{x \in \mathbf{R}^n \mid -1 \leq x_i \leq 1, \forall i\} \right\} \end{aligned} \quad (3.8.3)$$

The standard semidefinite relaxations of the problems are, respectively, the problems

$$\begin{aligned} (a) \quad & \max_X \{ \text{Tr}(AX) : X \succeq 0, X_{ii} = 1, i = 1, \dots, n \}, \\ (b) \quad & \max_X \{ \text{Tr}(AX) : X \succeq 0, X_{ii} \leq 1, i = 1, \dots, n \}; \end{aligned} \quad (3.8.4)$$

the optimal value of a relaxation is an upper bound for the optimal value of the respective original problem.

Exercise 3.30 *Let $A \in \mathbf{S}^n$. Prove that*

$$\max_{x: x_i = \pm 1, i=1,\dots,n} x^T A x \geq \text{Tr}(A).$$

Develop an efficient algorithm which, given A , generates a point x with coordinates ± 1 such that $x^T A x \geq \text{Tr}(A)$.

Exercise 3.31 *Prove that if the diagonal entries of A are nonnegative, then the optimal values in (3.8.4.a) and (3.8.4.b) are equal to each other. Thus, in the case in question, the relaxations "do not understand" whether we are maximizing over the vertices of the cube or over the entire cube.*

Exercise 3.32 *Prove that the problems dual to (3.8.4.a,b) are, respectively,*

$$\begin{aligned} (a) \quad & \min_{\Lambda} \{ \text{Tr}(\Lambda) : \Lambda \succeq A, \Lambda \text{ is diagonal} \}, \\ (b) \quad & \min_{\Lambda} \{ \text{Tr}(\Lambda) : \Lambda \succeq A, \Lambda \succeq 0, \Lambda \text{ is diagonal} \}; \end{aligned} \quad (3.8.5)$$

the optimal values in these problems are equal to those of the respective problems in (3.8.4) and are therefore upper bounds on the optimal values of the respective combinatorial problems from (3.8.3).

The latter claim is quite transparent, since the problems (3.8.5) can be obtained as follows:

- In order to bound from above the optimal value of a quadratic form $x^T Ax$ on a given set S , we look at those quadratic forms $x^T \Lambda x$ which can be easily maximized over S . For the case of $S = \text{Vrt}(C_n)$ these are quadratic forms with diagonal matrices Λ , and for the case of $S = C_n$ these are quadratic forms with diagonal and positive semidefinite matrices Λ ; in both cases, the respective maxima are merely $\text{Tr}(\Lambda)$.

- Having specified a family \mathcal{F} of quadratic forms $x^T \Lambda x$ “easily optimizable over S ”, we then look at those forms from \mathcal{F} which majorate everywhere the original quadratic form $x^T Ax$, and take among these forms the one with the minimal $\max_{x \in S} x^T \Lambda x$, thus coming to the problem

$$\min_{\Lambda} \left\{ \max_{x \in S} x^T \Lambda x : \Lambda \succeq A, \Lambda \in \mathcal{F} \right\}. \quad (!)$$

It is evident that the optimal value in this problem is an upper bound on $\max_{x \in S} x^T Ax$. It is also immediately seen that in the case of $S = \text{Vrt}(C_n)$ the problem $(!)$, with \mathcal{F} specified as the set \mathcal{D} of all diagonal matrices, is equivalent to (3.8.5.a), while in the case of $S = C_n$ $(!)$, with \mathcal{F} specified as the set \mathcal{D}_+ of positive semidefinite diagonal matrices, is nothing but (3.8.5.b).

Given the direct and quite transparent road leading to (3.8.5.a, b), we can try to move a little bit further along this road. To this end observe that there are trivial upper bounds on the maximum of an arbitrary quadratic form $x^T \Lambda x$ over $\text{Vrt}(C_n)$ and C_n , specifically:

$$\max_{x \in \text{Vrt}(C_n)} x^T \Lambda x \leq \text{Tr}(\Lambda) + \sum_{i \neq j} |\Lambda_{ij}|, \quad \max_{x \in C_n} x^T \Lambda x \leq \sum_{i,j} |\Lambda_{ij}|.$$

For the above families \mathcal{D} , \mathcal{D}_+ of matrices Λ for which $x^T \Lambda x$ is “easily optimizable” over $\text{Vrt}(C_n)$, respectively, C_n , the above bounds are equal to the precise values of the respective maxima. Now let us update $(!)$ as follows: we eliminate the restriction $\Lambda \in \mathcal{F}$, replacing simultaneously the objective $\max_{x \in S} x^T \Lambda x$ with its upper bound, thus coming to the pair of problems

$$(a) \quad \min_{\Lambda} \left\{ \text{Tr}(\Lambda) + \sum_{i \neq j} |\Lambda_{ij}| : \Lambda \succeq A \right\} \quad [S = \text{Vrt}(C_n)] \\ (b) \quad \min_{\Lambda} \left\{ \sum_{i,j} |\Lambda_{ij}| : \Lambda \succeq A \right\} \quad [S = C_n] \quad (3.8.6)$$

From the origin of the problems it is clear that they still yield upper bounds on the optimal values of the respective problems (3.8.3.a, b), and that these bounds are at least as good as the bounds yielded by the standard relaxations (3.8.4.a, b):

$$(a) \quad \text{Opt}(3.8.3.a) \leq \text{Opt}(3.8.6.a) \underbrace{\leq}_{(*)} \text{Opt}(3.8.4.a) = \text{Opt}(3.8.5.a), \\ (b) \quad \text{Opt}(3.8.3.b) \leq \text{Opt}(3.8.6.b) \underbrace{\leq}_{(**)} \text{Opt}(3.8.4.b) = \text{Opt}(3.8.5.b), \quad (3.8.7)$$

where $\text{Opt}(\cdot)$ means the optimal value of the corresponding problem.

Indeed, consider the problem (3.8.6.a). Whenever Λ is a feasible solution of this problem, the quadratic form $x^T \Lambda x$ majorates everywhere the form $x^T Ax$, so that $\max_{x \in \text{Vrt}(C_n)} x^T Ax \leq$

$\max_{x \in \text{Vrt}(C_n)} x^T \Lambda x$; the latter quantity, in turn, is majorated by $\text{Tr}(\Lambda) + \sum_{i \neq j} |\Lambda_{ij}|$, whence the value of the objective of the problem (3.8.6.a) at every feasible solution of the problem majorates the quantity $\max_{x \in \text{Vrt}(C_n)} x^T Ax$. Thus, the optimal value in (3.8.6.a) is an upper bound on the maximum of $x^T Ax$ over the vertices of the cube C_n . At the same time, when passing from the (dual form of the) standard relaxation (3.8.5.a) to our new bounding problem (3.8.6.a), we only extend the feasible set and do not vary the objective at the “old” feasible set; as a result of such a modification, the optimal value may only decrease. Thus, the upper bound on the maximum of $x^T Ax$ over $\text{Vrt}(C_n)$ yielded by (3.8.6.a) is at least as good as those (equal to each other) bounds yielded by the standard relaxations (3.8.4.a), (3.8.5.a), as required in (3.8.7.a). Similar reasoning proves (3.8.7.b).

Note that problems (3.8.6) are equivalent to semidefinite programs and thus are of the same status of “computational tractability” as the standard SDP relaxations (3.8.5) of the combinatorial problems in question. At the same time, our new bounding problems are more difficult than the standard SDP relaxations. Can we justify this by getting an improvement in the quality of the bounds?

Exercise 3.33 Find out whether the problems (3.8.6.a, b) yield better bounds than the respective problems (3.8.5.a, b), i.e., whether the inequalities (*), (**) in (3.8.7) can be strict.

Hint: Look at the problems dual to (3.8.6.a, b).

Exercise 3.34 Let D be a given subset of \mathbf{R}_+^n . Consider the following pair of optimization problems:

$$\begin{aligned} \max_x & \left\{ x^T Ax : (x_1^2, x_2^2, \dots, x_n^2)^T \in D \right\} && (P) \\ \max_X & \left\{ \text{Tr}(AX) : X \succeq 0, \text{Dg}(X) \in D \right\} && (R) \end{aligned}$$

($\text{Dg}(X)$ is the diagonal of a square matrix X). Note that when $D = \{(1, \dots, 1)^T\}$, (P) is the problem of maximizing a quadratic form over the vertices of C_n , while (R) is the standard semidefinite relaxation of (P); when $D = \{x \in \mathbf{R}^n \mid 0 \leq x_i \leq 1 \forall i\}$, (P) is the problem of maximizing a quadratic form over the cube C_n , and (R) is the standard semidefinite relaxation of the latter problem.

- 1) Prove that if D is semidefinite-representable, then (R) can be reformulated as a semidefinite program.
- 2) Prove that (R) is a relaxation of (P), i.e., that

$$\text{Opt}(P) \leq \text{Opt}(R).$$

- 3) [Nesterov; Ye] Let $A \succeq 0$. Prove that then

$$\text{Opt}(P) \leq \text{Opt}(R) \leq \frac{\pi}{2} \text{Opt}(P).$$

Hint: Use Nesterov’s Theorem (Theorem 3.4.2).

Exercise 3.35 Let $A \in \mathbf{S}_+^m$. Prove that

$$\max\{x^T Ax \mid x_i = \pm 1, i = 1, \dots, m\} = \max\left\{\frac{2}{\pi} \sum_{i,j=1}^m a_{ij} \text{asin}(X_{ij}) \mid X \succeq 0, X_{ii} = 1, i = 1, \dots, m\right\}.$$

3.8.4 Around Lyapunov Stability Analysis

A natural mathematical model of a swing is the linear time invariant dynamic system

$$y''(t) = -\omega^2 y(t) - 2\mu y'(t) \quad (\text{S})$$

with positive ω^2 and $0 \leq \mu < \omega$ (the term $2\mu y'(t)$ represents friction). A general solution to this equation is

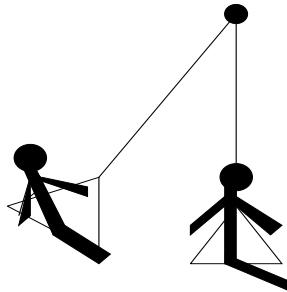
$$y(t) = a \cos(\omega' t + \phi_0) \exp\{-\mu t\}, \quad \omega' = \sqrt{\omega^2 - \mu^2},$$

with free parameters a and ϕ_0 , i.e., this is a decaying oscillation. Note that the equilibrium

$$y(t) \equiv 0$$

is stable – every solution to (S) converges to 0, along with its derivative, exponentially fast.

After stability is observed, an immediate question arises: how is it possible to swing on a swing? Everybody knows from practice that it is possible. On the other hand, since the equilibrium is stable, it looks as if it was impossible to swing, without somebody's assistance, for a long time. The reason which makes swinging possible is highly nontrivial – *parametric resonance*. A swinging child does not sit on the swing in a once forever fixed position; what he does is shown below:



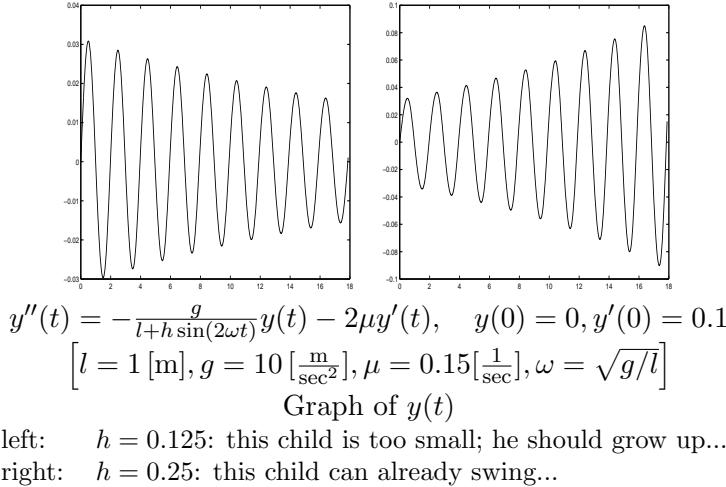
A swinging child

As a result, the “effective length” of the swing l – the distance from the point where the rope is fixed to the center of gravity of the system – is varying with time: $l = l(t)$. Basic mechanics says that $\omega^2 = g/l$, g being the gravity acceleration. Thus, the actual swing is a *time-varying* linear dynamic system:

$$y''(t) = -\omega^2(t)y(t) - 2\mu y'(t), \quad (\text{S}')$$

and it turns out that for properly varied $\omega(t)$ the equilibrium $y(t) \equiv 0$ is *not* stable. A swinging child is just varying $l(t)$ in a way which results in an *unstable* dynamic system (S'), and this

instability is in fact what the child enjoys...



Exercise 3.36 Assume that you are given parameters l (“nominal length of the swing rope”), $h > 0$ and $\mu > 0$, and it is known that a swinging child can vary the “effective length” of the rope within the bounds $l \pm h$, i.e., his/her movement is governed by the uncertain linear time-varying system

$$y''(t) = -a(t)y(t) - 2\mu y'(t), \quad a(t) \in \left[\frac{g}{l+h}, \frac{g}{l-h} \right].$$

Try to identify the domain in the 3D-space of parameters l, μ, h where the system is stable, as well as the domain where its stability can be certified by a quadratic Lyapunov function. What is “the difference” between these two domains?

3.8.5 Around Nesterov's $\frac{\pi}{2}$ Theorem

Exercise 3.37 Prove the following statement:

Proposition 3.8.1 [Nesterov;Ye] Consider the optimization program

$$\text{Opt} = \max_x \left\{ x^T A x : x^T B_i x \leq 1, i = 1, \dots, m \right\}, \quad (P)$$

along with its semidefinite relaxation

$$\text{SDP} = \max_X \left\{ \text{Tr}(AX) : \text{Tr}(B_i X) \leq b_i, i = 1, \dots, m, X \succeq 0 \right\} \quad (SDP)$$

and the dual of the latter problem:

$$\min_{\lambda} \left\{ \sum_i \lambda_i b_i : \sum_i \lambda_i B_i \succeq A, \lambda \geq 0 \right\}. \quad (SDD)$$

Assume that

1. The matrices B_1, \dots, B_m commute with each other;

2. There exists a combination of the matrices B_i with nonnegative coefficients which is positive definite;
3. $A \succeq 0$.

Then $\text{Opt} \geq 0$, (SDP) and (SDD) are solvable with equal optimal values, and

$$\text{Opt} \leq \text{SDP} \leq \frac{\pi}{2} \text{Opt}. \quad (3.8.8)$$

Hint: Observe that since B_i are commuting symmetric matrices, they share a common orthogonal eigenbasis, so that w.l.o.g. we can assume that all B_i 's are diagonal. In this latter case, use Theorem 3.4.3.

3.8.6 Around ellipsoidal approximations

Exercise 3.38 Prove the Löwner – Fritz John Theorem (Theorem 3.7.1).

More on ellipsoidal approximations of sums of ellipsoids. The goal of two subsequent exercises is to get in an alternative way the problem (O) “generating” a parametric family of ellipsoids containing the arithmetic sum of m given ellipsoids (Section 3.7.4).

Exercise 3.39 Let P_i be nonsingular, and Λ_i be positive definite $n \times n$ matrices, $i = 1, \dots, m$. Prove that for every collection x^1, \dots, x^m of vectors from \mathbf{R}^n one has

$$[x^1 + \dots + x^m]^T \left[\sum_{i=1}^m [P_i^T]^{-1} \Lambda_i^{-1} P_i^{-1} \right]^{-1} [x^1 + \dots + x^m] \leq \sum_{i=1}^m [x^i]^T P_i \Lambda_i P_i^T x^i. \quad (3.8.9)$$

Hint: Consider the $(nm + n) \times (nm + n)$ symmetric matrix

$$A = \begin{bmatrix} P_1 \Lambda_1 P_1^T & & I_n \\ & \ddots & \vdots \\ & & P_m \Lambda_m P_m^T & I_n \\ \hline I_n & \cdots & I_n & \sum_{i=1}^m [P_i^T]^{-1} \Lambda_i^{-1} P_i^{-1} \end{bmatrix}$$

and apply twice the Schur Complement Lemma: first time – to prove that the matrix is positive semidefinite, and the second time – to get from the latter fact the desired inequality.

Exercise 3.40 Assume you are given m full-dimensional ellipsoids centered at the origin

$$W_i = \{x \in \mathbf{R}^n \mid x^T B_i x \leq 1\}, \quad i = 1, \dots, m \quad [B_i \succ 0]$$

in \mathbf{R}^n .

- 1) Prove that for every collection Λ of positive definite $n \times n$ matrices Λ_i such that

$$\sum_i \lambda_{\max}(\Lambda_i) \leq 1$$

the ellipsoid

$$E_\Lambda = \{x \mid x^T \left[\sum_{i=1}^m B_i^{-1/2} \Lambda_i^{-1} B_i^{-1/2} \right]^{-1} x \leq 1\}$$

contains the sum $W_1 + \dots + W_m$ of the ellipsoids W_i .

2) Prove that in order to find the smallest volume ellipsoid in the family $\{E_\Lambda\}_\Lambda$ defined in 1) it suffices to solve the semidefinite program

$$\begin{aligned}
 (a) \quad & \text{maximize } t \text{ s.t.} \\
 (b) \quad & t \leq \text{Det}^{1/n}(Z), \\
 (c) \quad & Z \succeq 0, \\
 (d) \quad & \Lambda_i \preceq \lambda_i I_n, \quad i = 1, \dots, m, \\
 (e) \quad & \sum_{i=1}^m \lambda_i \leq 1
 \end{aligned} \tag{3.8.10}$$

in variables $Z, \Lambda_i \in \mathbf{S}^n, t, \lambda_i \in \mathbf{R}$; the smallest volume ellipsoid in the family $\{E_\Lambda\}_\Lambda$ is E_{Λ^*} , where Λ^* is the “ Λ -part” of an optimal solution of the problem.

Hint: Use example 20c from Lecture 3.

3) Demonstrate that the optimal value in (3.8.10) remains unchanged when the matrices Λ_i are further restricted to be scalar: $\Lambda_i = \lambda_i I_n$. Prove that with this additional constraint problem (3.8.10) becomes equivalent to problem $(\tilde{\mathcal{O}})$ from Section 3.7.4.

Remark 3.8.1 Exercise 3.40 demonstrates that the approximating scheme for solving problem (\mathcal{O}) presented in Proposition 3.7.4 is equivalent to the following one:

Given m positive reals λ_i with unit sum, one defines the ellipsoid $E(\lambda) = \{x \mid x^T \left[\sum_{i=1}^m \lambda_i^{-1} B_i^{-1} \right]^{-1} x \leq 1\}$. This ellipsoid contains the arithmetic sum W of the ellipsoids $\{x \mid x^T B_i x \leq 1\}$, and in order to approximate the smallest volume ellipsoid containing W , we merely minimize $\text{Det}(E(\lambda))$ over λ varying in the standard simplex $\{\lambda \geq 0, \sum_i \lambda_i = 1\}$.

In this form, the approximation scheme in question was proposed by Schweppe (1975).

Exercise 3.41 Let A_i be nonsingular $n \times n$ matrices, $i = 1, \dots, m$, and let $W_i = \{x = A_i u \mid u^T u \leq 1\}$ be the associated ellipsoids in \mathbf{R}^n . Let $\Delta_m = \{\lambda \in \mathbf{R}_+^m \mid \sum_i \lambda_i = 1\}$. Prove that

1) Whenever $\lambda \in \Delta_m$ and $A \in \mathbf{M}^{n,n}$ is such that

$$AA^T \succeq F(\lambda) \equiv \sum_{i=1}^m \lambda_i^{-1} A_i A_i^T,$$

the ellipsoid $E[A] = \{x = Au \mid u^T u \leq 1\}$ contains $W = W_1 + \dots + W_m$.

Hint: Use the result of Exercise 3.40.1)

2) Whenever $A \in \mathbf{M}^{n,n}$ is such that

$$AA^T \preceq F(\lambda) \quad \forall \lambda \in \Delta_m,$$

the ellipsoid $E[A]$ is contained in $W_1 + \dots + W_m$, and vice versa.

Hint: Note that

$$\left(\sum_{i=1}^m |\alpha_i| \right)^2 = \min_{\lambda \in \Delta_m} \sum_{i=1}^m \frac{\alpha_i^2}{\lambda_i}$$

and use statement **(F)** from Section 3.7.4.

“Simple” ellipsoidal approximations of sums of ellipsoids. Let $W_i = \{x = A_i u \mid u^T u \leq 1\}$, $i = 1, \dots, m$, be full-dimensional ellipsoids in \mathbf{R}^n (so that A_i are nonsingular $n \times n$ matrices), and let $W = W_1 + \dots + W_m$ be the arithmetic sum of these ellipsoids. Observe that W is the image of the set

$$\mathcal{B} = \left\{ u = \begin{bmatrix} u[1] \\ \vdots \\ u[m] \end{bmatrix} \in \mathbf{R}^{nm} \mid u^T[i]u[i] \leq 1, \quad i = 1, \dots, m \right\}$$

under the linear mapping

$$u \mapsto \mathcal{A}u = \sum_{i=1}^m A_i u[i] : \mathbf{R}^{nm} \rightarrow \mathbf{R}^n.$$

It follows that

Whenever an nm -dimensional ellipsoid \mathcal{W} contains \mathcal{B} , the set $\mathcal{A}(\mathcal{W})$, which is an n -dimensional ellipsoid (why?) contains W , and whenever \mathcal{W} is contained in \mathcal{B} , the ellipsoid $\mathcal{A}(\mathcal{W})$ is contained in W .

In view of this observation, we can try to approximate W from inside and from outside by the ellipsoids $W_- \equiv \mathcal{A}(\mathcal{W}_-)$ and $W^+ = \mathcal{A}(\mathcal{W}^+)$, where \mathcal{W}_- and \mathcal{W}^+ are, respectively, the largest and the smallest volume nm -dimensional ellipsoids contained in/containing \mathcal{B} .

Exercise 3.42 1) Prove that

$$\begin{aligned} \mathcal{W}_- &= \{u \in \mathbf{R}^{nm} \mid \sum_{i=1}^m u^T[i]u[i] \leq 1\}, \\ \mathcal{W}^+ &= \{u \in \mathbf{R}^{nm} \mid \sum_{i=1}^m u^T[i]u[i] \leq m\}, \end{aligned}$$

so that

$$\begin{aligned} W \supset W_- &\equiv \{x = \sum_{i=1}^m A_i u[i] \mid \sum_{i=1}^m u^T[i]u[i] \leq 1\}, \\ W \subset W_+ &\equiv \{x = \sum_{i=1}^m A_i u[i] \mid \sum_{i=1}^m u^T[i]u[i] \leq m\} = \sqrt{m}W_-. \end{aligned}$$

2) Prove that W_- can be represented as

$$W_- = \{x = Bu \mid u \in \mathbf{R}^n, u^T u \leq 1\}$$

with matrix $B \succ 0$ representable as

$$B = \sum_{i=1}^m A_i X_i$$

with square matrices X_i of norms $|X_i| \leq 1$.

Derive from this observation that the “level of conservativeness” of the inner ellipsoidal approximation of W given by Proposition 3.7.6 is at most \sqrt{m} : if W_* is this inner ellipsoidal approximation and W_{**} is the largest volume ellipsoid contained in W , then

$$\left(\frac{\text{Vol}(W_{**})}{\text{Vol}(W_*)} \right)^{1/n} \leq \left(\frac{\text{Vol}(W)}{\text{Vol}(W_*)} \right)^{1/n} \leq \sqrt{m}.$$

Invariant ellipsoids

Exercise 3.43 Consider a discrete time controlled dynamic system

$$\begin{aligned} x(t+1) &= Ax(t) + bu(t), \quad t = 0, 1, 2, \dots \\ x(0) &= 0, \end{aligned}$$

where $x(t) \in \mathbf{R}^n$ is the state vector and $u(t) \in [-1, 1]$ is the control at time t . An ellipsoid centered at the origin

$$W = \{x \mid x^T Z x \leq 1\} \quad [Z \succ 0]$$

is called invariant, if

$$x \in W \Rightarrow Ax \pm b \in W.$$

Prove that

- 1) If W is an invariant ellipsoid and $x(t) \in W$ for some t , then $x(t') \in W$ for all $t' \geq t$.
- 2) Assume that the vectors $b, Ab, A^2b, \dots, A^{n-1}b$ are linearly independent. Prove that an invariant ellipsoid exists if and only if A is stable (the absolute values of all eigenvalues of A are < 1).
- 3) Assuming that A is stable, prove that an ellipsoid $\{x \mid x^T Z x \leq 1\} [Z \succ 0]$ is invariant if and only if there exists $\lambda \geq 0$ such that

$$\begin{pmatrix} 1 - b^T Z b - \lambda & -b^T Z A \\ -A^T Z b & \lambda Z - A^T Z A \end{pmatrix} \succeq 0.$$

How could one use this fact to approximate numerically the smallest volume invariant ellipsoid?

Greedy “infinitesimal” ellipsoidal approximations. Consider a linear time-varying controlled system

$$\frac{d}{dt}x(t) = A(t)x(t) + B(t)u(t) + v(t) \quad (3.8.11)$$

with continuous matrix-valued functions $A(t)$, $B(t)$, continuous vector-valued function $v(\cdot)$ and norm-bounded control:

$$\|u(\cdot)\|_2 \leq 1. \quad (3.8.12)$$

Assume that the initial state of the system belongs to a given ellipsoid:

$$x(0) \in E(0) = \{x \mid (x - x^0)^T G^0(x - x^0) \leq 1\} \quad [G^0 = [G^0]^T \succ 0]. \quad (3.8.13)$$

Our goal is to build, in an “on-line” fashion, a system of ellipsoids

$$E(t) = \{x \mid (x - x_t)^T G_t(x - x_t) \leq 1\} \quad [G_t = G_t^T \succ 0] \quad (3.8.14)$$

in such a way that if $u(\cdot)$ is a control satisfying (3.8.12) and $x(0)$ is an initial state satisfying (3.8.13), then for every $t \geq 0$ it holds

$$x(t) \in E(t).$$

We are interested to minimize the volumes of the resulting ellipsoids.

There is no difficulty with the path x_t of centers of the ellipsoids: it “obviously” should satisfy the requirements

$$\frac{d}{dt}x_t = A(t)x_t + v(t), \quad t \geq 0; \quad x_0 = x^0. \quad (3.8.15)$$

Let us take this choice for granted and focus on how should we define the positive definite matrices G_t . Let us look for a continuously differentiable matrix-valued function G_t , taking values in the set of positive definite symmetric matrices, with the following property:

(L) For every $t \geq 0$ and every point $x^t \in E(t)$ (see (3.8.14)), every trajectory $x(\tau)$, $\tau \geq t$, of the system

$$\frac{d}{d\tau}x(\tau) = A(\tau)x(\tau) + B(\tau)u(\tau) + v(\tau), \quad x(t) = x^t$$

with $\|u(\cdot)\|_2 \leq 1$ satisfies $x(\tau) \in E(\tau)$ for all $\tau \geq t$.

Note that **(L)** is a sufficient (but in general not necessary) condition for the system of ellipsoids $E(t)$, $t \geq 0$, “to cover” all trajectories of (3.8.11) – (3.8.12). Indeed, when formulating **(L)**, we act as if we were sure that the states $x(t)$ of our system run through the entire ellipsoid $E(t)$, which is not necessarily the case. The advantage of **(L)** is that this condition can be converted into an “infinitesimal” form:

Exercise 3.44 Prove that if $G_t \succ 0$ is continuously differentiable and satisfies **(L)**, then

$$\forall \left(t \geq 0, x, u : x^T G_t x = 1, u^T u \leq 1 \right) : \quad 2u^T B^T(t)G_t x + x^T \left[\frac{d}{dt}G_t + A^T(t)G_t + G_t A(t) \right] x \leq 0. \quad (3.8.16)$$

Vice versa, if G_t is a continuously differentiable function taking values in the set of positive definite symmetric matrices and satisfying (3.8.16) and the initial condition $G_0 = G^0$, then the associated system of ellipsoids $\{E(t)\}$ satisfies **(L)**.

The result of Exercise 3.44 provides us with a kind of description of the families of ellipsoids $\{E(t)\}$ we are interested in. Now let us take care of the volumes of these ellipsoids. The latter can be done via a “greedy” (locally optimal) policy: given $E(t)$, let us try to minimize, under restriction (3.8.16), the derivative of the volume of the ellipsoid at time t . Note that this locally optimal policy does not necessary yield the smallest volume ellipsoids satisfying **(L)** (achieving “instant reward” is not always the best way to happiness); nevertheless, this policy makes sense.

We have $2 \ln \text{vol}(E_t) = -\ln \text{Det}(G_t)$, whence

$$2 \frac{d}{dt} \ln \text{vol}(E(t)) = -\text{Tr}(G_t^{-1} \frac{d}{dt} G_t);$$

thus, our greedy policy requires to choose $H_t \equiv \frac{d}{dt}G_t$ as a solution to the optimization problem

$$\max_{H=H^T} \left\{ \text{Tr}(G_t^{-1} H) : 2u^T B^T(t)G_t x + x^T \left[\frac{d}{dt}G_t - A^T(t)G_t - G_t A(t) \right] x \leq 0 \right. \\ \left. \forall \left(x, u : x^T G_t x = 1, u^T u \leq 1 \right) \right\}.$$

Exercise 3.45 Prove that the outlined greedy policy results in the solution G_t to the differential equation

$$\begin{aligned}\frac{d}{dt}G_t &= -A^T(t)G_t - G_tA(t) - \sqrt{\frac{n}{\text{Tr}(G_tB(t)B^T(t))}}G_tB(t)B^T(t)G_t - \sqrt{\frac{\text{Tr}(G_tB(t)B^T(t))}{n}}G_t, \quad t \geq 0; \\ G_0 &= G^0.\end{aligned}$$

Prove that the solution to this equation is symmetric and positive definite for all $t > 0$, provided that $G^0 = [G^0]^T \succ 0$.

Exercise 3.46 Modify the previous reasoning to demonstrate that the “locally optimal” policy for building inner ellipsoidal approximation of the set

$$\begin{aligned}X(t) &= \{x(t) \mid \exists x^0 \in E(0) \equiv \{x \mid (x - x^0)^T G^0 (x - x^0) \leq 1\}, \exists u(\cdot), \|u(\cdot)\|_2 \leq 1 : \\ &\quad \frac{d}{d\tau}x(\tau) = A(\tau)x(\tau) + B(\tau)u(\tau) + v(\tau), \quad 0 \leq \tau \leq t, \quad x(0) = x^0\}\end{aligned}$$

results in the family of ellipsoids

$$\underline{E}(t) = \{x \mid (x - x_t)^T W_t (x - x_t) \leq 1\},$$

where x_t is given by (3.8.15) and W_t is the solution of the differential equation

$$\frac{d}{dt}W_t = -A^T(t)W_t - W_tA(t) - 2W_t^{1/2}(W_t^{1/2}B(t)B^T(t)W_t^{1/2})^{1/2}W_t^{1/2}, \quad t \geq 0; \quad W_0 = G^0.$$

Lecture 4

Polynomial Time Interior Point algorithms for LP, CQP and SDP

4.1 Complexity of Convex Programming

When we attempt to solve any problem, we would like to know whether it is possible to find a correct solution in a “reasonable time”. Had we known that the solution will not be reached in the next 30 years, we would think (at least) twice before starting to solve it. Of course, this is an extreme case, but undoubtedly, it is highly desirable to distinguish between “computationally tractable” problems – those that can be solved efficiently – and problems which are “computationally intractable”. The corresponding *complexity theory* was first developed in Computer Science for combinatorial (discrete) problems, and later somehow extended onto the case of continuous computational problems, including those of Continuous Optimization. In this Section, we outline the main concepts of the CCT – Combinatorial Complexity Theory – along with their adaptations to Continuous Optimization.

4.1.1 Combinatorial Complexity Theory

A generic combinatorial problem is a family \mathcal{P} of *problem instances* of a “given structure”, each instance $(p) \in \mathcal{P}$ being identified by a finite-dimensional *data vector* $\text{Data}(p)$, specifying the particular values of the coefficients of “generic” analytic expressions. The data vectors are assumed to be *Boolean vectors* – with entries taking values 0, 1 only, so that the data vectors are, actually, finite binary words.

The model of computations in CCT: an idealized computer capable to store only integers (i.e., finite binary words), and its operations are *bitwise*: we are allowed to multiply, add and compare integers. To add and to compare two ℓ -bit integers, it takes $O(\ell)$ “bitwise” elementary operations, and to multiply a pair of ℓ -bit integers it costs $O(\ell^2)$ elementary operations (the cost of multiplication can be reduced to $O(\ell \ln(\ell))$, but it does not matter) .

In CCT, a solution to an instance (p) of a generic problem \mathcal{P} is a finite binary word y such that the pair $(\text{Data}(p), y)$ satisfies certain “verifiable condition” $\mathcal{A}(\cdot, \cdot)$. Namely, it is assumed that there exists a code \mathcal{M} for the above “Integer Arithmetic computer” such that executing the code on every input pair x, y of finite binary words, the computer after finitely many elementary

operations terminates and outputs either “yes”, if $\mathcal{A}(x, y)$ is satisfied, or “no”, if $\mathcal{A}(x, y)$ is not satisfied. Thus, \mathcal{P} is the problem

Given x , find y such that

$$\mathcal{A}(x, y) = \text{true}, \quad (4.1.1)$$

or detect that no such y exists.

For example, the problem *Stones*:

Given n positive integers a_1, \dots, a_n , find a vector $x = (x_1, \dots, x_n)^T$ with coordinates ± 1 such that $\sum_i x_i a_i = 0$, or detect that no such vector exists

is a generic combinatorial problem. Indeed, the data of the instance of the problem, same as candidate solutions to the instance, can be naturally encoded by finite sequences of integers. In turn, finite sequences of integers can be easily encoded by finite binary words. And, of course, for this problem you can easily point out a code for the “Integer Arithmetic computer” which, given on input two binary words $x = \text{Data}(p)$, y encoding the data vector of an instance (p) of the problem and a candidate solution, respectively, verifies in finitely many “bit” operations whether y represents or does not represent a solution to (p) .

A solution algorithm for a generic problem \mathcal{P} is a code \mathcal{S} for the Integer Arithmetic computer which, given on input the data vector $\text{Data}(p)$ of an instance $(p) \in \mathcal{P}$, after finitely many operations either returns a solution to the instance, or a (correct!) claim that no solution exists. The running time $T_{\mathcal{S}}(p)$ of the solution algorithm on instance (p) is exactly the number of elementary (i.e., bit) operations performed in course of executing \mathcal{S} on $\text{Data}(p)$.

A solvability test for a generic problem \mathcal{P} is defined similarly to a solution algorithm, but now all we want of the code is to say (correctly!) whether the input instance is or is not solvable, just “yes” or “no”, without constructing a solution in the case of the “yes” answer.

The complexity of a solution algorithm/solvability test \mathcal{S} is defined as

$$\text{Compl}_{\mathcal{S}}(\ell) = \max\{T_{\mathcal{S}}(p) \mid (p) \in \mathcal{P}, \text{length}(\text{Data}(p)) \leq \ell\},$$

where $\text{length}(x)$ is the bit length (i.e., number of bits) of a finite binary word x . The algorithm/test is called *polynomial time*, if its complexity is bounded from above by a polynomial of ℓ .

Finally, a generic problem \mathcal{P} is called to be *polynomially solvable*, if it admits a polynomial time solution algorithm. If \mathcal{P} admits a polynomial time solvability test, we say that \mathcal{P} is *polynomially verifiable*.

Classes P and NP. A generic problem \mathcal{P} is said to belong to the class NP, if the corresponding condition \mathcal{A} , see (4.1.1), possesses the following two properties:

I. \mathcal{A} is polynomially computable, i.e., the running time $T(x, y)$ (measured, of course, in elementary “bit” operations) of the associated code \mathcal{M} is bounded from above by a polynomial of the bit length $\text{length}(x) + \text{length}(y)$ of the input:

$$T(x, y) \leq \chi(\text{length}(x) + \text{length}(y))^\chi \quad \forall(x, y)^1$$

Thus, the first property of an NP problem states that *given the data Data(p) of a problem instance p and a candidate solution y , it is easy to check whether y is an actual solution of (p)* – to verify this fact, it suffices to compute $A(\text{Data}(p), y)$, and this computation requires polynomial in $\text{length}(\text{Data}(p)) + \text{length}(y)$ time.

The second property of an NP problem makes its instances even more easier:

II. A solution to an instance (p) of a problem cannot be “too long” as compared to the data of the instance: there exists χ such that

$$\text{length}(y) > \chi \text{length}^\chi(x) \Rightarrow \mathcal{A}(x, y) = \text{"no"}$$

A generic problem \mathcal{P} is said to belong to the class P, if it belongs to the class NP and is polynomially solvable.

NP-completeness is defined as follows:

Definition 4.1.1 (i) Let \mathcal{P}, \mathcal{Q} be two problems from NP. Problem \mathcal{Q} is called to be *polynomially reducible to \mathcal{P}* , if there exists a polynomial time algorithm \mathcal{M} (i.e., a code for the Integer Arithmetic computer with the running time bounded by a polynomial of the length of the input) with the following property. Given on input the data vector $\text{Data}(q)$ of an instance $(q) \in \mathcal{Q}$, \mathcal{M} converts this data vector to the data vector $\text{Data}(p[q])$ of an instance of \mathcal{P} such that $(p[q])$ is solvable if and only if (q) is solvable.

(ii) A generic problem \mathcal{P} from NP is called *NP-complete*, if every other problem \mathcal{Q} from NP is polynomially reducible to \mathcal{P} .

The importance of the notion of an NP-complete problem comes from the following fact:

If a particular NP-complete problem is polynomially verifiable (i.e., admits a polynomial time solvability test), then every problem from NP is polynomially solvable: $P = NP$.

The question whether $P=NP$ – whether NP-complete problems are or are not polynomially solvable, is qualified as “the most important open problem in Theoretical Computer Science” and remains open for about 30 years. One of the most basic results of Theoretical Computer Science is that NP-complete problems do exist (the Stones problem is an example). Many of these problems are of huge practical importance, and are therefore subject, over decades, of intensive studies of thousands excellent researchers. However, no polynomial time algorithm for any of these problems was found. Given the huge total effort invested in this research, we should conclude that it is “highly improbable” that NP-complete problems are polynomially solvable. Thus, at the “practical level” the fact that certain problem is NP-complete is sufficient to qualify the problem as “computationally intractable”, at least at our present level of knowledge.

¹⁾Here and in what follows, we denote by χ positive “characteristic constants” associated with the predicates/problems in question. The particular values of these constants are of no importance, the only thing that matters is their existence. Note that in different places of even the same equation χ may have different values.

4.1.2 Complexity in Continuous Optimization

It is convenient to represent continuous optimization problems as *Mathematical Programming problems*, i.e. programs of the following form:

$$\min_x \{p_0(x) : x \in X(p) \subset \mathbf{R}^{n(p)}\} \quad (p)$$

where

- $n(p)$ is the *design dimension* of program (p) ;
- $X(p) \subset \mathbf{R}^n$ is the *feasible domain* of the program;
- $p_0(x) : \mathbf{R}^n \rightarrow \mathbf{R}$ is the *objective* of (p) .

Families of optimization programs. We want to speak about methods for solving optimization programs (p) “of a given structure” (for example, Linear Programming ones). All programs (p) “of a given structure”, like in the combinatorial case, form certain family \mathcal{P} , and we assume that every particular program in this family – every *instance* (p) of \mathcal{P} – is specified by its particular *data* $\text{Data}(p)$. However, now the data is a finite-dimensional *real* vector; one may think about the entries of this data vector as about particular values of coefficients of “generic” (specific for \mathcal{P}) analytic expressions for $p_0(x)$ and $X(p)$. The dimension of the vector $\text{Data}(p)$ will be called the *size* of the instance:

$$\text{Size}(p) = \dim \text{Data}(p).$$

The model of computations. This is what is known as “Real Arithmetic Model of Computations”, as opposed to “Integer Arithmetic Model” in the CCT. We assume that the computations are carried out by an idealized version of the usual computer which is capable to store countably many reals and can perform with them the standard *exact* real arithmetic operations – the four basic arithmetic operations, evaluating elementary functions, like cos and exp, and making comparisons.

Accuracy of approximate solutions. We assume that a generic optimization problem \mathcal{P} is equipped with an “infeasibility measure” $\text{Infeas}_{\mathcal{P}}(x, p)$ – a real-valued function of $p \in \mathcal{P}$ and $x \in \mathbf{R}^{n(p)}$ which quantifies the infeasibility of vector x as a candidate solution to (p) . In our general considerations, all we require from this measure is that

- $\text{Infeas}_{\mathcal{P}}(x, p) \geq 0$, and $\text{Infeas}_{\mathcal{P}}(x, p) = 0$ when x is feasible for (p) (i.e., when $x \in X(p)$).

Given an infeasibility measure, we can proceed to define the notion of an ϵ -solution to an instance $(p) \in \mathcal{P}$, namely, as follows. Let

$$\text{Opt}(p) \in \{-\infty\} \cup \mathbf{R} \cup \{+\infty\}$$

be the optimal value of the instance (i.e., the infimum of the values of the objective on the feasible set, if the instance is feasible, and $+\infty$ otherwise). A point $x \in \mathbf{R}^{n(p)}$ is called an ϵ -solution to (p) , if

$$\text{Infeas}_{\mathcal{P}}(x, p) \leq \epsilon \text{ and } p_0(x) - \text{Opt}(p) \leq \epsilon,$$

i.e., if x is both “ ϵ -feasible” and “ ϵ -optimal” for the instance.

It is convenient to define the *number of accuracy digits* in an ϵ -solution to (p) as the quantity

$$\text{Digits}(p, \epsilon) = \ln \left(\frac{\text{Size}(p) + \|\text{Data}(p)\|_1 + \epsilon^2}{\epsilon} \right).$$

Solution methods. A solution method \mathcal{M} for a given family \mathcal{P} of optimization programs is a code for the idealized Real Arithmetic computer. When solving an instance $(p) \in \mathcal{P}$, the computer first inputs the data vector $\text{Data}(p)$ of the instance and a real $\epsilon > 0$ – the accuracy to which the instance should be solved, and then executes the code \mathcal{M} on this input. We assume that the execution, on every input $(\text{Data}(p), \epsilon > 0)$ with $(p) \in \mathcal{P}$, takes finitely many elementary operations of the computer, let this number be denoted by $\text{Compl}_{\mathcal{M}}(p, \epsilon)$, and results in one of the following three possible outputs:

- an $n(p)$ -dimensional vector $\text{Res}_{\mathcal{M}}(p, \epsilon)$ which is an ϵ -solution to (p) ,
- a correct message “ (p) is infeasible”,
- a correct message “ (p) is unbounded below”.

We measure the efficiency of a method by its *running time* $\text{Compl}_{\mathcal{M}}(p, \epsilon)$ – the number of elementary operations performed by the method when solving instance (p) within accuracy ϵ . By definition, the fact that \mathcal{M} is “efficient” (*polynomial time*) on \mathcal{P} , means that there exists a polynomial $\pi(s, \tau)$ such that

$$\begin{aligned} \text{Compl}_{\mathcal{M}}(p, \epsilon) &\leq \pi(\text{Size}(p), \text{Digits}(p, \epsilon)) \\ \forall (p) \in \mathcal{P} \quad \forall \epsilon > 0. \end{aligned} \tag{4.1.2}$$

Informally speaking, polynomiality of \mathcal{M} means that when we increase the size of an instance and the required number of accuracy digits by absolute constant factors, the running time increases by no more than another absolute constant factor.

We call a family \mathcal{P} of optimization problems *polynomially solvable* (or, which is the same, *computationally tractable*), if it admits a polynomial time solution method.

4.1.3 Computational tractability of convex optimization problems

A generic optimization problem \mathcal{P} is called *convex*, if, for every instance $(p) \in \mathcal{P}$, both the objective $p_0(x)$ of the instance and the infeasibility measure $\text{Infeas}_{\mathcal{P}}(x, p)$ are convex functions of $x \in \mathbf{R}^{n(p)}$. One of the major complexity results in Continuous Optimization is that a generic *convex optimization problem, under mild computability and regularity assumptions, is polynomially solvable* (and thus “computationally tractable”). To formulate the precise result, we start with specifying the aforementioned “mild assumptions”.

Polynomial computability. Let \mathcal{P} be a generic convex program, and let $\text{Infeas}_{\mathcal{P}}(x, p)$ be the corresponding measure of infeasibility of candidate solutions. We say that our family is *polynomially computable*, if there exist two codes \mathcal{C}_{obj} , $\mathcal{C}_{\text{cons}}$ for the Real Arithmetic computer such that

1. For every instance $(p) \in \mathcal{P}$, the computer, when given on input the data vector of the instance (p) and a point $x \in \mathbf{R}^{n(p)}$ and executing the code \mathcal{C}_{obj} , outputs the value $p_0(x)$ and a subgradient $e(x) \in \partial p_0(x)$ of the objective p_0 of the instance at the point x , and the running time (i.e., total number of operations) of this computation $T_{\text{obj}}(x, p)$ is bounded from above by a polynomial of the size of the instance:

$$\forall ((p) \in \mathcal{P}, x \in \mathbf{R}^{n(p)}) : \quad T_{\text{obj}}(x, p) \leq \chi \text{Size}^{\chi}(p) \quad [\text{Size}(p) = \dim \text{Data}(p)]. \tag{4.1.3}$$

(recall that in our notation, χ is a common name of characteristic constants associated with \mathcal{P}).

2. For every instance $(p) \in \mathcal{P}$, the computer, when given on input the data vector of the instance (p) , a point $x \in \mathbf{R}^{n(p)}$ and an $\epsilon > 0$ and executing the code $\mathcal{C}_{\text{cons}}$, reports on output whether $\text{Infeas}_{\mathcal{P}}(x, p) \leq \epsilon$, and if it is not the case, outputs a linear form a which separates the point x from all those points y where $\text{Infeas}_{\mathcal{P}}(y, p) \leq \epsilon$:

$$\forall (y, \text{Infeas}_{\mathcal{P}}(y, p) \leq \epsilon) : \quad a^T x > a^T y, \quad (4.1.4)$$

the running time $T_{\text{cons}}(x, \epsilon, p)$ of the computation being bounded by a polynomial of the size of the instance and of the “number of accuracy digits”:

$$\forall ((p) \in \mathcal{P}, x \in \mathbf{R}^{n(p)}, \epsilon > 0) : \quad T_{\text{cons}}(x, \epsilon, p) \leq \chi (\text{Size}(p) + \text{Digits}(p, \epsilon))^{\chi}. \quad (4.1.5)$$

Note that the vector a in (4.1.4) is not supposed to be nonzero; when it is 0, (4.1.4) simply says that there are no points y with $\text{Infeas}_{\mathcal{P}}(y, p) \leq \epsilon$.

Polynomial growth. We say that a generic convex program \mathcal{P} is with *polynomial growth*, if the objectives and the infeasibility measures, as functions of x , grow polynomially with $\|x\|_1$, the degree of the polynomial being a power of $\text{Size}(p)$:

$$\begin{aligned} \forall ((p) \in \mathcal{P}, x \in \mathbf{R}^{n(p)}) : \\ |p_0(x)| + \text{Infeas}_{\mathcal{P}}(x, p) \leq (\chi [\text{Size}(p) + \|x\|_1 + \|\text{Data}(p)\|_1])^{\chi \text{Size}^{\chi}(p)}. \end{aligned} \quad (4.1.6)$$

Polynomial boundedness of feasible sets. We say that a generic convex program \mathcal{P} has *polynomially bounded feasible sets*, if the feasible set $X(p)$ of every instance $(p) \in \mathcal{P}$ is bounded, and is contained in the Euclidean ball, centered at the origin, of “not too large” radius:

$$\begin{aligned} \forall (p) \in \mathcal{P} : \\ X(p) \subset \left\{ x \in \mathbf{R}^{n(p)} : \|x\|_2 \leq (\chi [\text{Size}(p) + \|\text{Data}(p)\|_1])^{\chi \text{Size}^{\chi}(p)} \right\}. \end{aligned} \quad (4.1.7)$$

Example. Consider generic optimization problems \mathcal{LP}_b , \mathcal{CQP}_b , \mathcal{SDP}_b with instances in the conic form

$$\min_{x \in \mathbf{R}^{n(p)}} \left\{ p_0(x) \equiv c_{(p)}^T x : x \in X(p) \equiv \{x : A_{(p)}x - b_{(p)} \in \mathbf{K}(p), \|x\|_2 \leq R\} \right\}; \quad (4.1.8)$$

where \mathbf{K} is a cone belonging to a characteristic for the generic program family \mathcal{K} of cones, specifically,

- the family of nonnegative orthants for \mathcal{LP}_b ,
- the family of *direct products* of Lorentz cones for \mathcal{CQP}_b ,
- the family of semidefinite cones for \mathcal{SDP}_b .

The *data* of an instance (p) of the type (4.1.8) is the collection

$$\text{Data}(p) = (n(p), c_{(p)}, A_{(p)}, b_{(p)}, R, \langle \text{size}(s) \text{ of } \mathbf{K}_{(p)} \rangle),$$

with naturally defined *size(s)* of a cone \mathbf{K} from the family \mathcal{K} associated with the generic program under consideration: the sizes of \mathbf{R}_+^n and of \mathbf{S}_+^n equal n , and the size of a direct product of Lorentz cones is the sequence of the dimensions of the factors.

The generic conic programs in question are equipped with the infeasibility measure

$$\text{Infeas}(x, p) = \min \left\{ t \geq 0 : t\mathbf{e}[\mathbf{K}_{(p)}] + A_{(p)}x - b_{(p)} \in \mathbf{K}_{(p)} \right\}, \quad (4.1.9)$$

where $\mathbf{e}[\mathbf{K}]$ is a naturally defined “central point” of $\mathbf{K} \in \mathcal{K}$, specifically,

- the n -dimensional vector of ones when $\mathbf{K} = \mathbf{R}_+^n$,
- the vector $\mathbf{e}_m = (0, \dots, 0, 1)^T \in \mathbf{R}^m$ when $\mathbf{K}_{(p)}$ is the Lorentz cone \mathbf{L}^m , and the direct sum of these vectors, when \mathbf{K} is a direct product of Lorentz cones,
- the unit matrix of appropriate size when \mathbf{K} is a semidefinite cone.

In the sequel, we refer to the three generic problems we have just defined as to *Linear, Conic Quadratic and Semidefinite Programming problems with ball constraints*, respectively. It is immediately seen that the generic programs \mathcal{LP}_b , \mathcal{CQP}_b and \mathcal{SDP}_b are convex and possess the properties of polynomial computability, polynomial growth and polynomially bounded feasible sets (the latter property is ensured by making the ball constraint $\|x\|_2 \leq R$ a part of program’s formulation).

Computational Tractability of Convex Programming. The role of the properties we have introduced becomes clear from the following result:

Theorem 4.1.1 *Let \mathcal{P} be a family of convex optimization programs equipped with infeasibility measure $\text{Infeas}_{\mathcal{P}}(\cdot, \cdot)$. Assume that the family is polynomially computable, with polynomial growth and with polynomially bounded feasible sets. Then \mathcal{P} is polynomially solvable.*

In particular, the generic Linear, Conic Quadratic and Semidefinite programs with ball constraints \mathcal{LP}_b , \mathcal{CQP}_b , \mathcal{SDP}_b are polynomially solvable.

4.1.4 What is inside Theorem 4.1.1: Black-box represented convex programs and the Ellipsoid method

Theorem 4.1.1 is a more or less straightforward corollary of a result related to the so called *Information-Based complexity of black-box represented convex programs*. This result is interesting by its own right, this is why we reproduce it here:

Consider a Convex Programming program

$$\min_x \{f(x) : x \in X\} \quad (4.1.10)$$

where

- X is a convex compact set in \mathbf{R}^n with a nonempty interior
- f is a continuous convex function on X .

Assume that our “environment” when solving (4.1.10) is as follows:

1. We have access to a *Separation Oracle* $\text{Sep}(X)$ for X – a routine which, given on input a point $x \in \mathbf{R}^n$, reports on output whether or not $x \in \text{int } X$, and in the case of $x \notin \text{int } X$, returns a *separator* – a nonzero vector e such that

$$e^T x \geq \max_{y \in X} e^T y \quad (4.1.11)$$

(the existence of such a separator is guaranteed by the Separation Theorem for convex sets);

2. We have access to a *First Order* oracle which, given on input a point $x \in \text{int } X$, returns the value $f(x)$ and a subgradient $f'(x)$ of f at x (Recall that a subgradient $f'(x)$ of f at x is a vector such that

$$f(y) \geq f(x) + (y - x)^T f'(x) \quad (4.1.12)$$

for all y ; convex function possesses subgradients at every relative interior point of its domain, see Section C.6.2.);

3. We are given two positive reals $R \geq r$ such that X is contained in the Euclidean ball, centered at the origin, of the radius R and contains a Euclidean ball of the radius r (not necessarily centered at the origin).

The result we are interested in is as follows:

Theorem 4.1.2 *In the outlined “working environment”, for every given $\epsilon > 0$ it is possible to find an ϵ -solution to (4.1.10), i.e., a point $x_\epsilon \in X$ with*

$$f(x_\epsilon) \leq \min_{x \in X} f(x) + \epsilon$$

in no more than $N(\epsilon)$ subsequent calls to the Separation and the First Order oracles plus no more than $O(1)n^2N(\epsilon)$ arithmetic operations to process the answers of the oracles, with

$$N(\epsilon) = O(1)n^2 \ln \left(2 + \frac{\text{Var}_X(f)R}{\epsilon \cdot r} \right). \quad (4.1.13)$$

Here

$$\text{Var}_X(f) = \max_X f - \min_X f.$$

4.1.4.1 Proof of Theorem 4.1.2: the Ellipsoid method

Assume that we are interested to solve the convex program (4.1.10) and we have an access to a separation oracle $\text{Sep}(X)$ for the feasible domain of (4.1.10) and to a first order oracle $\mathcal{O}(f)$ for the objective of (4.1.10). How could we solve the problem via these “tools”? An extremely transparent way is given by the *Ellipsoid method* which can be viewed as a multi-dimensional extension of the usual bisection.

Ellipsoid method: the idea. Assume that we have already found an n -dimensional ellipsoid

$$E = \{x = c + Bu \mid u^T u \leq 1\} \quad [B \in \mathbf{M}^{n,n}, \text{Det}B \neq 0]$$

which contains the optimal set X_* of (4.1.10) (note that $X_* \neq \emptyset$, since the feasible set X of (4.1.10) is assumed to be compact, and the objective f – to be convex on the entire \mathbf{R}^n and therefore continuous, see Theorem C.4.1). How could we construct a smaller ellipsoid containing X_* ?

The answer is immediate.

1) Let us call the separation oracle $\text{Sep}(X)$, the center c of the current ellipsoid being the input. There are two possible cases:

1.a) $\text{Sep}(X)$ reports that $c \notin X$ and returns a separator a :

$$a \neq 0, \quad a^T c \geq \sup_{y \in X} a^T y. \quad (4.1.14)$$

In this case we can replace our current “localizer” E of the optimal set X_* by a smaller one – namely, by the “half-ellipsoid”

$$\widehat{E} = \{x \in E \mid a^T x \leq a^T c\}.$$

Indeed, by assumption $X_* \subset E$; when passing from E to \widehat{E} , we cut off all points x of E where $a^T x > a^T c$, and by (4.1.14) all these points are outside of X and therefore outside of $X_* \subset X$. Thus, $X_* \subset \widehat{E}$.

1.b) $\text{Sep}(X)$ reports that $c \in X$. In this case we call the first order oracle $\mathcal{O}(f)$, c being the input; the oracle returns the value $f(c)$ and a subgradient $a = f'(c)$ of f at c . Again, two cases are possible:

1.b.1) $a = 0$. In this case we are done – c is a minimizer of f on X . Indeed, $c \in X$, and (4.1.12) reads

$$f(y) \geq f(c) + 0^T(y - c) = f(c) \quad \forall y \in \mathbf{R}^n.$$

Thus, c is a minimizer of f on \mathbf{R}^n , and since $c \in X$, c minimizes f on X as well.

1.b.2) $a \neq 0$. In this case (4.1.12) reads

$$a^T(x - c) > 0 \Rightarrow f(x) > f(c),$$

so that replacing the ellipsoid E with the half-ellipsoid

$$\widehat{E} = \{x \in E \mid a^T x \leq a^T c\}$$

we ensure the inclusion $X_* \subset \widehat{E}$. Indeed, $X_* \subset E$ by assumption, and when passing from E to \widehat{E} , we cut off all points of E where $a^T x > a^T c$ and, consequently, where $f(x) > f(c)$; since $c \in X$, no one of these points can belong to the set X_* of minimizers of f on X .

2) We have seen that as a result of operations described in 1.a-b) we either terminate with an exact minimizer of f on X , or obtain a “half-ellipsoid”

$$\widehat{E} = \{x \in E \mid a^T x \leq a^T c\} \quad [a \neq 0]$$

containing X_* . It remains to use the following simple geometric fact:

(*) Let $E = \{x = c + Bu \mid u^T u \leq 1\}$ ($\text{Det}B \neq 0$) be an n -dimensional ellipsoid and $\widehat{E} = \{x \in E \mid a^T x \leq a^T c\}$ ($a \neq 0$) be a “half” of E . If $n > 1$, then \widehat{E} is contained in the ellipsoid

$$\begin{aligned} E^+ &= \{x = c^+ + B^+u \mid u^T u \leq 1\}, \\ c^+ &= c - \frac{1}{n+1}Bp, \\ B^+ &= B \left(\frac{n}{\sqrt{n^2-1}}(I_n - pp^T) + \frac{n}{n+1}pp^T \right) = \frac{n}{\sqrt{n^2-1}}B + \left(\frac{n}{n+1} - \frac{n}{\sqrt{n^2-1}} \right)(Bp)p^T, \\ p &= \frac{B^T a}{\sqrt{a^T B B^T a}} \end{aligned} \tag{4.1.15}$$

and if $n = 1$, then the set \widehat{E} is contained in the ellipsoid (which now is just a segment)

$$\begin{aligned} E^+ &= \{x \mid c^+ B^+ u \mid |u| \leq 1\}, \\ c^+ &= c - \frac{1}{2} \frac{Ba}{|Ba|}, \\ B_+ &= \frac{1}{2}B. \end{aligned}$$

In all cases, the n -dimensional volume $\text{Vol}(E^+)$ of the ellipsoid E^+ is less than the one of E :

$$\text{Vol}(E^+) = \left(\frac{n}{\sqrt{n^2 - 1}} \right)^{n-1} \frac{n}{n+1} \text{Vol}(E) \leq \exp\{-1/(2n)\} \text{Vol}(E) \quad (4.1.16)$$

$$(\text{in the case of } n = 1, \left(\frac{n}{\sqrt{n^2 - 1}} \right)^{n-1} = 1).$$

(*) says that there exists (and can be explicitly specified) an ellipsoid $E^+ \supset \widehat{E}$ with the volume constant times less than the one of E . Since E^+ covers \widehat{E} , and the latter set, as we have seen, covers X_* , E^+ covers X_* . Now we can iterate the above construction, thus obtaining a sequence of ellipsoids $E, E^+, (E^+)^+, \dots$ with volumes going to 0 at a linear rate (depending on the dimension n only) which “collapses” to the set X_* of optimal solutions of our problem – exactly as in the usual bisection!

Note that (*) is just an exercise in elementary calculus. Indeed, the ellipsoid E is given as an image of the unit Euclidean ball $W = \{u \mid u^T u \leq 1\}$ under the one-to-one affine mapping $u \mapsto c + Bu$; the half-ellipsoid \widehat{E} is then the image, under the same mapping, of the half-ball

$$\widehat{W} = \{u \in W \mid p^T u \leq 0\}$$

p being the unit vector from (4.1.15); indeed, if $x = c + Bu$, then $a^T x \leq a^T c$ if and only if $a^T Bu \leq 0$, or, which is the same, if and only if $p^T u \leq 0$. Now, instead of covering \widehat{E} by a small in volume ellipsoid E^+ , we may cover by a small ellipsoid W^+ the half-ball \widehat{W} and then take E^+ to be the image of W^+ under our affine mapping:

$$E^+ = \{x = c + Bu \mid u \in W^+\}. \quad (4.1.17)$$

Indeed, if W^+ contains \widehat{W} , then the image of W^+ under our affine mapping $u \mapsto c + Bu$ contains the image of \widehat{W} , i.e., contains \widehat{E} . And since the ratio of volumes of two bodies remain invariant under affine mapping (passing from a body to its image under an affine mapping $u \mapsto c + Bu$, we just multiply the volume by $|\text{Det}B|$), we have

$$\frac{\text{Vol}(E^+)}{\text{Vol}(E)} = \frac{\text{Vol}(W^+)}{\text{Vol}(W)}.$$

Thus, the problem of finding a “small” ellipsoid E^+ containing the half-ellipsoid \widehat{E} can be reduced to the one of finding a “small” ellipsoid W^+ containing the half-ball \widehat{W} , as shown on Fig. 5.1. Now, the problem of finding a small ellipsoid containing \widehat{W} is very simple: our “geometric data” are invariant with respect to rotations around the p -axis, so that we may look for W^+ possessing the same rotational symmetry. Such an ellipsoid W^+ is given by just 3 parameters: its center should belong to our symmetry axis, i.e., should be $-hp$ for certain h , one of the half-axes of the ellipsoid (let its length be μ) should be directed along p , and the remaining $n - 1$ half-axes should be of the same length λ and be orthogonal to p . For our 3 parameters h, μ, λ we have 2 equations expressing the fact that the boundary of W^+ should pass through the “South pole” $-p$ of W and trough the “equator” $\{u \mid u^T u = 1, p^T u = 0\}$; indeed, W^+ should contain \widehat{W} and thus – both the pole and the equator, and since we are looking for W^+ with the smallest possible volume, both the pole and the equator should be on the boundary of W^+ . Using our 2 equations to express μ and λ via h , we end up with a single “free” parameter h , and the volume of W^+ (i.e., $\text{const}(n)\mu\lambda^{n-1}$) becomes an explicit function of h ; minimizing this function in h , we find the “optimal” ellipsoid W^+ , check that it indeed contains \widehat{W} (i.e., that our geometric intuition was correct) and then convert W^+ into E^+ according to (4.1.17), thus coming to the explicit formulas (4.1.15) –

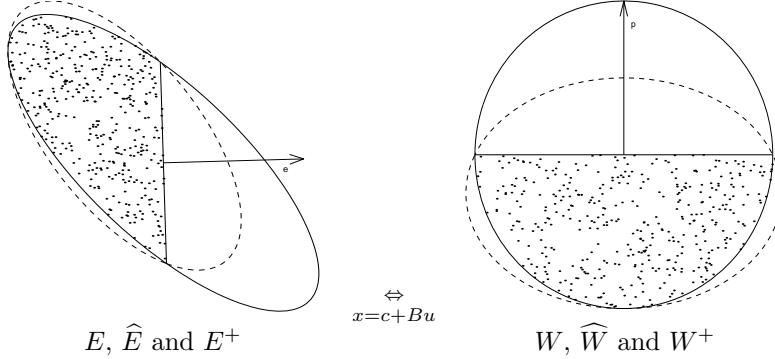


Figure 5.1.

(4.1.16); implementation of the outlined scheme takes from 10 to 30 minutes, depending on how many miscalculations are made...

It should be mentioned that although the indicated scheme is quite straightforward and elementary, the fact that it works is not evident a priori: it might happen that the smallest volume ellipsoid containing a half-ball is just the original ball! This would be the death of our idea – instead of a sequence of ellipsoids collapsing to the solution set X_* , we would get a “stationary” sequence $E, E, E\dots$ Fortunately, it is not happening, and this is a great favour Nature does to Convex Optimization...

Ellipsoid method: the construction. There is a small problem with implementing our idea of “trapping” the optimal set X_* of (4.1.10) by a “collapsing” sequence of ellipsoids. The only thing we can ensure is that all our ellipsoids contain X_* and that their volumes rapidly (at a linear rate) converge to 0. However, the linear sizes of the ellipsoids should not necessarily go to 0 – it may happen that the ellipsoids are shrinking in some directions and are not shrinking (or even become larger) in other directions (look what happens if we apply the construction to minimizing a function of 2 variables which in fact depends only on the first coordinate). Thus, to the moment it is unclear how to build a sequence of points converging to X_* . This difficulty, however, can be easily resolved: as we shall see, we can form this sequence from the best feasible solutions generated so far. Another issue which remains open to the moment is when to terminate the method; as we shall see in a while, this issue also can be settled satisfactory.

The precise description of the Ellipsoid method as applied to (4.1.10) is as follows (in this description, we assume that $n \geq 2$, which of course does not restrict generality):

The Ellipsoid Method.

Initialization. Recall that when formulating (4.1.10) it was assumed that the feasible set X of the problem is contained in the ball $E_0 = \{x \mid \|x\|_2 \leq R\}$ of a given radius R and contains an (unknown) Euclidean ball of a known radius $r > 0$. The ball E_0 will be our initial ellipsoid; thus, we set

$$c_0 = 0, \quad B_0 = RI, \quad E_0 = \{x = c_0 + B_0 u \mid u^T u \leq 1\};$$

note that $E_0 \supset X$.

We also set

$$\rho_0 = R, \quad L_0 = 0.$$

The quantities ρ_t will be the “radii” of the ellipsoids E_t to be built, i.e., the radii of the Euclidean balls of the same volumes as E_t ’s. The quantities L_t will be our guesses for the variation

$$\text{Var}_R(f) = \max_{x \in E_0} f(x) - \min_{x \in E_0} f(x)$$

of the objective on the initial ellipsoid E_0 . We shall use these guesses in the termination test.

Finally, we input the accuracy $\epsilon > 0$ to which we want to solve the problem.

Step t , $t = 1, 2, \dots$. At the beginning of step t , we have the previous ellipsoid

$$E_{t-1} = \{x = c_{t-1} + B_{t-1}u \mid u^T u \leq 1\} \quad [c_{t-1} \in \mathbf{R}^n, B_{t-1} \in \mathbf{M}^{n,n}, \text{Det}B_{t-1} \neq 0]$$

(i.e., have c_{t-1} , B_{t-1}) along with the quantities $L_{t-1} \geq 0$ and

$$\rho_{t-1} = |\text{Det}B_{t-1}|^{1/n}.$$

At step t , we act as follows (cf. the preliminary description of the method):

1) We call the separation oracle $\text{Sep}(X)$, c_{t-1} being the input. It is possible that the oracle reports that $c_{t-1} \notin X$ and provides us with a separator

$$a \neq 0 : \quad a^T c_{t-1} \geq \sup_{y \in X} a^T y.$$

In this case we call step t *non-productive*, set

$$a_t = a, \quad L_t = L_{t-1}$$

and go to rule 3) below. Otherwise – i.e., when $c_{t-1} \in X$ – we call step t *productive* and go to rule 2).

2) We call the first order oracle $\mathcal{O}(f)$, c_{t-1} being the input, and get the value $f(c_{t-1})$ and a subgradient $a \equiv f'(c_{t-1})$ of f at the point c_{t-1} . It is possible that $a = 0$; in this case we terminate and claim that c_{t-1} is an optimal solution to (4.1.10). In the case of $a \neq 0$ we set

$$a_t = a,$$

compute the quantity

$$\ell_t = \max_{y \in E_0} [a_t^T y - a_t^T c_{t-1}] = R \|a_t\|_2 - a_t^T c_{t-1},$$

update L by setting

$$L_t = \max\{L_{t-1}, \ell_t\}$$

and go to rule 3).

3) We set

$$\hat{E}_t = \{x \in E_{t-1} \mid a_t^T x \leq a_t^T c_{t-1}\}$$

(cf. the definition of \hat{E} in our preliminary description of the method) and define the new ellipsoid

$$E_t = \{x = c_t + B_t u \mid u^T u \leq 1\}$$

by setting (see (4.1.15))

$$\begin{aligned} p_t &= \frac{B_{t-1}^T a_t}{\sqrt{a_t^T B_{t-1} B_{t-1}^T a_t}} \\ c_t &= c_{t-1} - \frac{1}{n+1} B_{t-1} p_t, \\ B_t &= \frac{n}{\sqrt{n^2-1}} B_{t-1} + \left(\frac{n}{n+1} - \frac{n}{\sqrt{n^2-1}} \right) (B_{t-1} p_t) p_t^T. \end{aligned} \quad (4.1.18)$$

We also set

$$\rho_t = |\text{Det} B_t|^{1/n} = \left(\frac{n}{\sqrt{n^2-1}} \right)^{(n-1)/n} \left(\frac{n}{n+1} \right)^{1/n} \rho_{t-1}$$

(see (4.1.16)) and go to rule 4).

4) [Termination test]. We check whether the inequality

$$\frac{\rho_t}{r} < \frac{\epsilon}{L_t + \epsilon} \quad (4.1.19)$$

is satisfied. If it is the case, we terminate and output, as the result of the solution process, the best (i.e., with the smallest value of f) of the “search points” $c_{\tau-1}$ associated with productive steps $\tau \leq t$ (we shall see that these productive steps indeed exist, so that the result of the solution process is well-defined). If (4.1.19) is not satisfied, we go to step $t+1$.

Just to get some feeling how the method works, here is a 2D illustration. The problem is

$$\begin{aligned} f(x) &= \min_{-1 \leq x_1, x_2 \leq 1} f(x), \\ &\quad \frac{1}{2}(1.443508244x_1 + 0.623233851x_2 - 7.957418455)^2 \\ &\quad + 5(-0.350974738x_1 + 0.799048618x_2 + 2.877831823)^4, \end{aligned}$$

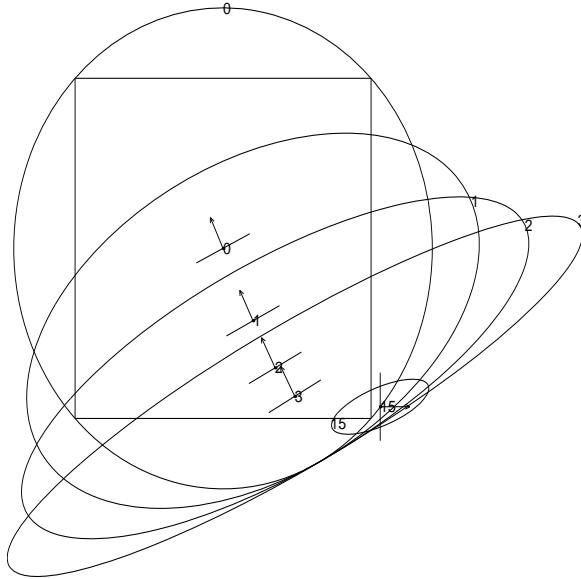
the optimal solution is $x_1^* = 1, x_2^* = -1$, and the exact optimal value is 70.030152768...

The values of f at the best (i.e., with the smallest value of the objective) feasible solutions found in course of first t steps of the method, $t = 1, 2, \dots, 256$, are shown in the following table:

t	best value	t	best value
1	374.61091739	16	76.838253451
2	216.53084103
3	146.74723394	32	70.901344815
4	112.42945457
5	93.84206347	64	70.031633483
6	82.90928589
7	82.90928589	128	70.030154192
8	82.90928589
...	...	256	70.030152768

The initial phase of the process looks as shown on Fig. 5.2.

Ellipsoid method: complexity analysis. We are about to establish our key result (which, in particular, immediately implies Theorem 4.1.2):

Figure 5.2. Ellipses E_{t-1} and search points c_{t-1} , $t = 1, 2, 3, 4, 16$

Arrows: gradients of the objective $f(x)$
 Unmarked segments: tangents to the level lines of $f(x)$

Theorem 4.1.3 Let the Ellipsoid method be applied to convex program (4.1.10) of dimension $n \geq 2$ such that the feasible set X of the problem contains a Euclidean ball of a given radius $r > 0$ and is contained in the ball $E_0 = \{\|x\|_2 \leq R\}$ of a given radius R . For every input accuracy $\epsilon > 0$, the Ellipsoid method terminates after no more than

$$N(\epsilon) = \text{Ceil} \left(2n^2 \left[\ln \left(\frac{R}{r} \right) + \ln \left(\frac{\epsilon + \text{Var}_R(f)}{\epsilon} \right) \right] \right) + 1 \quad (4.1.20)$$

steps, where

$$\text{Var}_R(f) = \max_{E_0} f - \min_{E_0} f,$$

and $\text{Ceil}(a)$ is the smallest integer $\geq a$. Moreover, the result \hat{x} generated by the method is a feasible ϵ -solution to (4.1.10):

$$\hat{x} \in X \text{ and } f(x) - \min_X f \leq \epsilon. \quad (4.1.21)$$

Proof. We should prove the following pair of statements:

- (i) The method terminates in course of the first $N(\epsilon)$ steps
- (ii) The result \hat{x} is a feasible ϵ -solution to the problem.

¹⁰. Comparing the preliminary and the final description of the Ellipsoid method and taking into account the initialization rule, we see that if the method does not terminate before step t or terminates at this step according to rule 4), then

$$\begin{aligned}
 (a) \quad & E_0 \supset X; \\
 (b) \quad & E_\tau \supset \widehat{E}_\tau = \left\{ x \in E_{\tau-1} \mid a_\tau^T x \leq a_\tau^T c_{\tau-1} \right\}, \quad \tau = 1, \dots, t; \\
 (c) \quad \text{Vol}(E_\tau) &= \rho_\tau^n \text{Vol}(E_0) = \left(\frac{n}{\sqrt{n^2-1}} \right)^{n-1} \frac{n}{n+1} \text{Vol}(E_{\tau-1}) \\
 &\leq \exp\{-1/(2n)\} \text{vol}(E_{\tau-1}), \quad \tau = 1, \dots, t.
 \end{aligned} \tag{4.1.22}$$

Note that from (c) it follows that

$$\rho_\tau \leq \exp\{-\tau/(2n^2)\}R, \quad \tau = 1, \dots, t. \quad (4.1.23)$$

2⁰. We claim that

If the Ellipsoids method terminates at certain step t , then the result \hat{x} is well-defined and is a feasible ϵ -solution to (4.1.10).

Indeed, there are only two possible reasons for termination. First, it may happen that $c_{t-1} \in X$ and $f'(c_{t-1}) = 0$ (see rule 2)). From our preliminary considerations we know that in this case c_{t-1} is an optimal solution to (4.1.10), which is even more than what we have claimed. Second, it may happen that at step t relation (4.1.19) is satisfied. Let us prove that the claim of 2⁰ takes place in this case as well.

2⁰.a) Let us set

$$\nu = \frac{\epsilon}{\epsilon + L_t} \in (0, 1].$$

By (4.1.19), we have $\rho_t/r < \nu$, so that there exists ν' such that

$$\frac{\rho_t}{r} < \nu' < \nu \quad [\leq 1]. \quad (4.1.24)$$

Let x_* be an optimal solution to (4.1.10), and X^+ be the “ ν' -shrinkage” of X to x_* :

$$X^+ = x_* + \nu'(X - x_*) = \{x = (1 - \nu')x_* + \nu'z \mid z \in X\}. \quad (4.1.25)$$

We have

$$\text{Vol}(X^+) = (\nu')^n \text{Vol}(X) \geq \left(\frac{r\nu'}{R}\right)^n \text{Vol}(E_0) \quad (4.1.26)$$

(the last inequality is given by the fact that X contains a Euclidean ball of the radius r), while

$$\text{Vol}(E_t) = \left(\frac{\rho_t}{R}\right)^n \text{Vol}(E_0) \quad (4.1.27)$$

by definition of ρ_t . Comparing (4.1.26), (4.1.27) and taking into account that $\rho_t < r\nu'$ by (4.1.24), we conclude that $\text{Vol}(E_t) < \text{Vol}(X^+)$ and, consequently, X^+ cannot be contained in E_t . Thus, there exists a point y which belongs to X^+ :

$$y = (1 - \nu')x_* + \nu'z \quad [z \in X], \quad (4.1.28)$$

and does not belong to E_t .

2⁰.b) Since y does not belong to E_t and at the same time belongs to $X \subset E_0$ along with x_* and z (X is convex!), we see that there exists a $\tau \leq t$ such that $y \in E_{\tau-1}$ and $y \notin E_\tau$. By (4.1.22.b), every point x from the complement of E_τ in $E_{\tau-1}$ satisfies the relation $a_\tau^T x > a_\tau^T c_{\tau-1}$. Thus, we have

$$a_\tau^T y > a_\tau^T c_{\tau-1} \quad (4.1.29)$$

2⁰.c) Observe that the step τ is surely productive. Indeed, otherwise, by construction of the method, a_τ would separate X from $c_{\tau-1}$, and (4.1.29) would be impossible (we know that $y \in X$!). Notice that in particular we have just proved that if the method terminates at a step t , then at least one of the steps $1, \dots, t$ is productive, so that the result is well-defined.

Since step τ is productive, a_τ is a subgradient of f at $c_{\tau-1}$ (description of the method!), so that

$$f(u) \geq f(c_{\tau-1}) + a_\tau^T(u - c_{\tau-1})$$

for all $u \in X$, and in particular for $u = x_*$. On the other hand, $z \in X \subset E_0$, so that by the definition of ℓ_τ and L_τ we have

$$a_\tau^T(z - c_{\tau-1}) \leq \ell_\tau \leq L_\tau.$$

Thus,

$$\begin{aligned} f(x_*) &\geq f(c_{\tau-1}) + a_\tau^T(x_* - c_{\tau-1}) \\ L_\tau &\geq a_\tau^T(z - c_{\tau-1}) \end{aligned}$$

Multiplying the first inequality by $(1 - \nu')$, the second – by ν' and adding the results, we get

$$\begin{aligned} (1 - \nu')f(x_*) + \nu'L_\tau &\geq (1 - \nu')f(c_{\tau-1}) + a_\tau^T([(1 - \nu')x_* + \nu'z] - c_{\tau-1}) \\ &= (1 - \nu')f(c_{\tau-1}) + a_\tau^T(y - c_{\tau-1}) \\ &\quad [\text{see (4.1.28)}] \\ &\geq (1 - \nu')f(c_{\tau-1}) \\ &\quad [\text{see (4.1.29)}] \end{aligned}$$

and we come to

$$\begin{aligned} f(c_{\tau-1}) &\leq f(x_*) + \frac{\nu'L_\tau}{1-\nu'} \\ &\leq f(x_*) + \frac{\nu'L_t}{1-\nu'} \\ &\quad [\text{since } L_\tau \leq L_t \text{ in view of } \tau \leq t] \\ &\leq f(x_*) + \epsilon \\ &\quad [\text{by definition of } \nu \text{ and since } \nu' < \nu] \\ &= \text{Opt}(C) + \epsilon. \end{aligned}$$

We see that there exists a productive (i.e., with feasible $c_{\tau-1}$) step $\tau \leq t$ such that the corresponding search point $c_{\tau-1}$ is ϵ -optimal. Since we are in the situation where the result \hat{x} is the best of the feasible search points generated in course of the first t steps, \hat{x} is also feasible and ϵ -optimal, as claimed in 2⁰.

³⁰ It remains to verify that the method does terminate in course of the first $N = N(\epsilon)$ steps. Assume, on the contrary, that it is not the case, and let us lead this assumption to a contradiction.

First, observe that for every productive step t we have

$$c_{t-1} \in X \text{ and } a_t = f'(c_{t-1}),$$

whence, by the definition of a subgradient and the variation $\text{Var}_R(f)$,

$$u \in E_0 \Rightarrow \text{Var}_R(f) \geq f(u) - f(c_{t-1}) \geq a_t^T(u - c_{t-1}),$$

whence

$$\ell_t \equiv \max_{u \in E_0} a_t^T(u - c_{t-1}) \leq \text{Var}_R(f).$$

Looking at the description of the method, we conclude that

$$L_t \leq \text{Var}_R(f) \quad \forall t. \tag{4.1.30}$$

Since we have assumed that the method does not terminate in course of the first N steps, we have

$$\frac{\rho_N}{r} \geq \frac{\epsilon}{\epsilon + L_N}. \quad (4.1.31)$$

The right hand side in this inequality is $\geq \epsilon/(\epsilon + \text{Var}_R(f))$ by (4.1.30), while the left hand side is $\leq \exp\{-N/(2n^2)\}R$ by (4.1.23). We get

$$\exp\{-N/(2n^2)\}R/r \geq \frac{\epsilon}{\epsilon + \text{Var}_R(f)} \Rightarrow N \leq 2n^2 \left[\ln\left(\frac{R}{r}\right) + \ln\left(\frac{\epsilon + \text{Var}_R(f)}{\epsilon}\right) \right],$$

which is the desired contradiction (see the definition of $N = N(\epsilon)$ in (4.1.20)). ■

4.1.5 Difficult continuous optimization problems

Real Arithmetic Complexity Theory can borrow from the Combinatorial Complexity Theory techniques for detecting “computationally intractable” problems. Consider the situation as follows: we are given a family \mathcal{P} of optimization programs and want to understand whether the family is computationally tractable. An affirmative answer can be obtained from Theorem 4.1.1; but how could we justify that the family is *intractable*? A natural course of action here is to demonstrate that certain difficult (NP-complete) combinatorial problem \mathcal{Q} can be reduced to \mathcal{P} in such a way that the possibility to solve \mathcal{P} in polynomial time would imply similar possibility for \mathcal{Q} . Assume that the objectives of the instances of \mathcal{P} are polynomially computable, and that we can point out a generic combinatorial problem \mathcal{Q} known to be NP-complete which can be reduced to \mathcal{P} in the following sense:

There exists a CCT-polynomial time algorithm \mathcal{M} which, given on input the data vector $\text{Data}(q)$ of an instance $(q) \in \mathcal{Q}$, converts it into a triple $\text{Data}(p[q]), \epsilon(q), \mu(q)$ comprised of the data vector of an instance $(p[q]) \in \mathcal{P}$, positive rational $\epsilon(q)$ and rational $\mu(q)$ such that $(p[q])$ is solvable and

- if (q) is unsolvable, then the value of the objective of $(p[q])$ at every $\epsilon(q)$ -solution to this problem is $\leq \mu(q) - \epsilon(q)$;
- if (q) is solvable, then the value of the objective of $(p[q])$ at every $\epsilon(q)$ -solution to this problem is $\geq \mu(q) + \epsilon(q)$.

We claim that in the case in question we have all reasons to qualify \mathcal{P} as a “computationally intractable” problem. Assume, on contrary, that \mathcal{P} admits a polynomial time solution method \mathcal{S} , and let us look what happens if we apply this algorithm to solve $(p[q])$ within accuracy $\epsilon(q)$. Since $(p[q])$ is solvable, the method must produce an $\epsilon(q)$ -solution \hat{x} to $(p[q])$. With additional “polynomial time effort” we may compute the value of the objective of $(p[q])$ at \hat{x} (recall that the objectives of instances from \mathcal{P} are assumed to be polynomially computable). Now we can compare the resulting value of the objective with $\mu(q)$; by definition of reducibility, if this value is $\leq \mu(q)$, q is unsolvable, otherwise q is solvable. Thus, we get a correct “Real Arithmetic” solvability test for \mathcal{Q} . By definition of a Real Arithmetic polynomial time algorithm, the running time of the test is bounded by a polynomial of $s(q) = \text{Size}(p[q])$ and of the quantity

$$d(q) = \text{Digits}((p[q]), \epsilon(q)) = \ln\left(\frac{\text{Size}(p[q]) + \|\text{Data}(p[q])\|_1 + \epsilon^2(q)}{\epsilon(q)}\right).$$

Now note that if $\ell = \text{length}(\text{Data}(q))$, then the total number of bits in $\text{Data}(p[q])$ and in $\epsilon(q)$ is bounded by a polynomial of ℓ (since the transformation $\text{Data}(q) \mapsto (\text{Data}(p[q]), \epsilon(q), \mu(q))$ takes

CCT-polynomial time). It follows that both $s(q)$ and $d(q)$ are bounded by polynomials in ℓ , so that our “Real Arithmetic” solvability test for \mathcal{Q} takes polynomial in $\text{length}(\text{Data}(q))$ number of arithmetic operations.

Recall that \mathcal{Q} was assumed to be an NP-complete generic problem, so that it would be “highly improbable” to find a polynomial time solvability test for this problem, while we have managed to build such a test. We conclude that the polynomial solvability of \mathcal{P} is highly improbable as well.

4.2 Interior Point Polynomial Time Methods for LP, CQP and SDP

4.2.1 Motivation

Theorem 4.1.1 states that generic convex programs, under mild computability and boundedness assumptions, are polynomially solvable. This result is extremely important theoretically; however, from the practical viewpoint it is, essentially, no more than “an existence theorem”. Indeed, the “universal” complexity bounds coming from Theorem 4.1.2, although polynomial, are not that attractive: by Theorem 4.1.1, when solving problem (4.1.10) with n design variables, the “price” of an accuracy digit (what it costs to reduce current inaccuracy ϵ by factor 2) is $O(n^2)$ calls to the first order and the separation oracles plus $O(n^4)$ arithmetic operations to process the answers of the oracles. Thus, even for simplest objectives to be minimized over simplest feasible sets, the arithmetic price of an accuracy digit is $O(n^4)$; think how long will it take to solve a problem with, say, 1,000 variables (which is still a “small” size for many applications). The good news about the methods underlying Theorem 4.1.2 is their universality: all they need is a Separation oracle for the feasible set and the possibility to compute the objective and its subgradient at a given point, which is not that much. The bad news about these methods has the same source as the good news: the methods are “oracle-oriented” and capable to use only *local* information on the program they are solving, in contrast to the fact that when solving instances of well-structured programs, like LP, we from the very beginning have in our disposal complete *global* description of the instance. And of course it is ridiculous to use a *complete global* knowledge of the instance just to mimic the *local* in their nature first order and separation oracles. What we would like to have is an optimization technique capable to “utilize efficiently” our global knowledge of the instance and thus allowing to get a solution much faster than it is possible for “nearly blind” oracle-oriented algorithms. The major event in the “recent history” of Convex Optimization, called sometimes “Interior Point revolution”, was the invention of these “smart” techniques.

4.2.2 Interior Point methods

The Interior Point revolution was started by the seminal work of N. Karmarkar (1984) where the first interior point method for LP was proposed; in 18 years since then, interior point (IP) polynomial time methods have become an extremely deep and rich theoretically and highly promising computationally area of Convex Optimization. A somehow detailed overview of the history and the recent state of this area is beyond the scope of this course; an interested reader is referred to [33, 35, 24] and references therein. All we intend to do is to give an idea of what are the IP methods, skipping nearly all (sometimes highly instructive and nontrivial) technicalities.

The simplest way to get a proper impression of the (most of) IP methods is to start with a quite traditional *interior penalty* scheme for solving optimization problems.

4.2.2.1 The Newton method and the Interior penalty scheme

Unconstrained minimization and the Newton method. Seemingly the simplest convex optimization problem is the one of unconstrained minimization of a smooth strongly convex objective:

$$\min_x \{f(x) : x \in \mathbf{R}^n\}; \quad (\text{UC})$$

a “smooth strongly convex” in this context means a 3 times continuously differentiable convex function f such that $f(x) \rightarrow \infty$, $\|x\|_2 \rightarrow \infty$, and such that the Hessian matrix $f''(x) = [\frac{\partial^2 f(x)}{\partial x_i \partial x_j}]$ of f is positive definite at every point x . Among numerous techniques for solving (UC), the most remarkable one is the Newton method. In its pure form, the Newton method is extremely transparent and natural: given a current iterate x , we approximate our objective f by its second-order Taylor expansion at the iterate – by the quadratic function

$$f_x(y) = f(x) + (y - x)^T f'(x) + \frac{1}{2}(y - x)^T f''(x)(y - x)$$

– and choose as the next iterate x_+ the minimizer of this quadratic approximation. Thus, the Newton method merely iterates the updating

$$x \mapsto x_+ = x - [f''(x)]^{-1} f'(x). \quad (\text{Nwt})$$

In the case of a (strongly convex) quadratic objective, the approximation coincides with the objective itself, so that the method reaches the exact solution in one step. It is natural to guess (and indeed is true) that in the case when the objective is smooth and strongly convex (although not necessarily quadratic) and the current iterate x is close enough to the minimizer x_* of f , the next iterate x_+ , although not being x_* exactly, will be “much closer” to the exact minimizer than x . The precise (and easy) result is that the Newton method converges *locally quadratically*, i.e., that

$$\|x_+ - x_*\|_2 \leq C \|x - x_*\|_2^2,$$

provided that $\|x - x_*\|_2 \leq r$ with small enough value of $r > 0$ (both this value and C depend on f). Quadratic convergence means essentially that eventually every new step of the process increases by a constant factor the number of accuracy digits in the approximate solution.

When started not “close enough” to the minimizer, the “pure” Newton method (Nwt) can demonstrate weird behaviour (look, e.g., what happens when the method is applied to the univariate function $f(x) = \sqrt{1 + x^2}$). The simplest way to overcome this drawback is to pass from the pure Newton method to its *damped* version

$$x \mapsto x_+ = x - \gamma(x) [f''(x)]^{-1} f'(x), \quad (\text{NwtD})$$

where the stepsize $\gamma(x) > 0$ is chosen in a way which, on one hand, ensures global convergence of the method and, on the other hand, enforces $\gamma(x) \rightarrow 1$ as $x \rightarrow x_*$, thus ensuring fast (essentially the same as for the pure Newton method) asymptotic convergence of the process²⁾.

²⁾ There are many ways to provide the required behaviour of $\gamma(x)$, e.g., to choose $\gamma(x)$ by a linesearch in the direction $e(x) = -[f''(x)]^{-1} f'(x)$ of the Newton step:

$$\gamma(x) = \operatorname{argmin}_t f(x + t e(x)).$$

Practitioners thought the (properly modified) Newton method to be the fastest, in terms of the iteration count, routine for smooth (not necessarily convex) unconstrained minimization, although sometimes “too heavy” for practical use: the practical drawbacks of the method are both the necessity to invert the Hessian matrix at each step, which is computationally costly in the large-scale case, and especially the necessity to compute this matrix (think how difficult it is to write a code computing 5,050 second order derivatives of a messy function of 100 variables).

Classical interior penalty scheme: the construction. Now consider a constrained convex optimization program. As we remember, one can w.l.o.g. make its objective linear, moving, if necessary, the actual objective to the list of constraints. Thus, let the problem be

$$\min_x \left\{ c^T x : x \in \mathcal{X} \subset \mathbf{R}^n \right\}, \quad (\text{C})$$

where \mathcal{X} is a closed convex set, which we assume to possess a nonempty interior. How could we solve the problem?

Traditionally it was thought that the problems of smooth convex unconstrained minimization are “easy”; thus, a quite natural desire was to reduce the constrained problem (C) to a series of smooth *unconstrained* optimization programs. To this end, let us choose somehow a *barrier* (another name – “an interior penalty function”) $F(x)$ for the feasible set \mathcal{X} – a function which is well-defined (and is smooth and strongly convex) on the interior of \mathcal{X} and “blows up” as a point from $\text{int } \mathcal{X}$ approaches a boundary point of \mathcal{X} :

$$x_i \in \text{int } \mathcal{X}, \quad x \equiv \lim_{i \rightarrow \infty} x_i \in \partial \mathcal{X} \Rightarrow F(x_i) \rightarrow \infty, \quad i \rightarrow \infty,$$

and let us look at the one-parametric family of functions generated by our objective and the barrier:

$$F_t(x) = tc^T x + F(x) : \text{int } \mathcal{X} \rightarrow \mathbf{R}.$$

Here the *penalty parameter* t is assumed to be nonnegative.

It is easily seen that under mild regularity assumptions (e.g., in the case of bounded \mathcal{X} , which we assume from now on)

- Every function $F_t(\cdot)$ attains its minimum over the interior of \mathcal{X} , the minimizer $x_*(t)$ being unique;
- The *central path* $x_*(t)$ is a smooth curve, and all its limiting, $t \rightarrow \infty$, points belong to the set of optimal solutions of (C).

This fact is quite clear intuitively. To minimize $F_t(\cdot)$ for large t is the same as to minimize the function $f_\rho(x) = c^T x + \rho F(x)$ for small $\rho = \frac{1}{t}$. When ρ is small, the function f_ρ is very close to $c^T x$ everywhere in \mathcal{X} , except a narrow stripe along the boundary of \mathcal{X} , the stripe becoming thinner and thinner as $\rho \rightarrow 0$. Therefore we have all reasons to believe that the minimizer of F_t for large t (i.e., the minimizer of f_ρ for small ρ) must be close to the set of minimizers of $c^T x$ on \mathcal{X} .

We see that the central path $x_*(t)$ is a kind of Ariadne’s thread which leads to the solution set of (C). On the other hand, to reach, given a value $t \geq 0$ of the penalty parameter, the point $x_*(t)$ on this path is the same as to minimize a smooth strongly convex function $F_t(\cdot)$ which attains its minimum at an interior point of \mathcal{X} . The latter problem is “nearly unconstrained one”, up to the fact that its objective is not everywhere defined. However, we can easily adapt the methods of unconstrained minimization, including the Newton one, to handle “nearly unconstrained”

problems. We see that constrained convex optimization in a sense can be reduced to the “easy” unconstrained one. The conceptually simplest way to make use of this observation would be to choose a “very large” value \bar{t} of the penalty parameter, like $\bar{t} = 10^6$ or $\bar{t} = 10^{10}$, and to run an unconstrained minimization routine, say, the Newton method, on the function $F_{\bar{t}}$, thus getting a good approximate solution to (C) “in one shot”. This policy, however, is impractical: since we have no idea where $x_*(\bar{t})$ is, we normally will start our process of minimizing $F_{\bar{t}}$ very far from the minimizer of this function, and thus for a long time will be unable to exploit fast local convergence of the method for unconstrained minimization we have chosen. A smarter way to use our Ariadne’s thread is exactly the one used by Theseus: to follow the thread. Assume, e.g., that we know in advance the minimizer of $F_0 \equiv F$, i.e., the point $x_*(0)$ ³⁾. Thus, we know where the central path starts. Now let us follow this path: at i -th step, standing at a point x_i “close enough” to some point $x_*(t_i)$ of the path, we

- first, increase a bit the current value t_i of the penalty parameter, thus getting a new “target point” $x_*(t_{i+1})$ on the path,
- and
- second, approach our new target point $x_*(t_{i+1})$ by running, say, the Newton method, started at our current iterate x_i , on the function $F_{t_{i+1}}$, until a new iterate x_{i+1} “close enough” to $x_*(t_{i+1})$ is generated.

As a result of such a step, we restore the initial situation – we again stand at a point which is close to a point on the central path, but this latter point has been moved along the central path towards the optimal set of (C). Iterating this updating and strengthening appropriately our “close enough” requirements as the process goes on, we, same as the central path, approach the optimal set. A conceptual advantage of this “path-following” policy as compared to the “brute force” attempt to reach a target point $x_*(\bar{t})$ with large \bar{t} is that now we have a hope to exploit all the time the strongest feature of our “working horse” (the Newton method) – its fast local convergence. Indeed, assuming that x_i is close to $x_*(t_i)$ and that we do not increase the penalty parameter too rapidly, so that $x_*(t_{i+1})$ is close to $x_*(t_i)$ (recall that the central path is smooth!), we conclude that x_i is close to our new target point $x_*(t_i)$. If all our “close enough” and “not too rapidly” are properly controlled, we may ensure x_i to be in the domain of the quadratic convergence of the Newton method as applied to $F_{t_{i+1}}$, and then it will take a quite small number of steps of the method to recover closeness to our new target point.

Classical interior penalty scheme: the drawbacks. At a qualitative “common sense” level, the interior penalty scheme looks quite attractive and extremely flexible: for the majority of optimization problems treated by the classical optimization, there is a plenty of ways to build a relatively simple barrier meeting all the requirements imposed by the scheme, there is a huge room to play with the policies for increasing the penalty parameter and controlling closeness to the central path, etc. And the theory says that under quite mild and general assumptions on the choice of the numerous “free parameters” of our construction, it still is guaranteed to converge to the optimal set of the problem we have to solve. All looks wonderful, until we realize that the convergence ensured by the theory is completely “unqualified”, it is a purely asymptotical phenomenon: we are promised to reach eventually a solution of a whatever accuracy we wish, but how long it will take for a given accuracy – this is the question the “classical” optimization

³⁾ There is no difficulty to ensure thus assumption: given an arbitrary barrier F and an arbitrary starting point $\bar{x} \in \text{int } \mathcal{X}$, we can pass from F to a new barrier $\bar{F} = F(x) - (x - \bar{x})^T F'(\bar{x})$ which attains its minimum exactly at \bar{x} , and then use the new barrier \bar{F} instead of our original barrier F ; and for the traditional approach we are following for the time being, F has absolutely no advantages as compared to \bar{F} .

theory, with its “convergence” – “asymptotic linear/superlinear/quadratic convergence” neither posed nor answered. And since our life in this world is finite (moreover, usually more finite than we would like it to be), “asymptotical promises” are perhaps better than nothing, but definitely are not all we would like to know. What is vitally important for us in theory (and to some extent – also in practice) is the issue of *complexity*: given an instance of such and such generic optimization problem and a desired accuracy ϵ , how large is the computational effort (# of arithmetic operations) needed to get an ϵ -solution of the instance? And we would like the answer to be a kind of a polynomial time complexity bound, and not a quantity depending on “unobservable and uncontrollable” properties of the instance, like the “level of regularity” of the boundary of \mathcal{X} at the (unknown!) optimal solution of the instance.

It turns out that the intuitively nice classical theory we have outlined is unable to say a single word on the complexity issues (it is how it should be: a reasoning in purely qualitative terms like “smooth”, “strongly convex”, etc., definitely cannot yield a quantitative result...) Moreover, from the complexity viewpoint just the very philosophy of the classical convex optimization turns out to be wrong:

- As far as the complexity is concerned, for nearly all “black box represented” classes of unconstrained convex optimization problems (those where all we know is that the objective is called $f(x)$, is (strongly) convex and 2 (3,4,5...) times continuously differentiable, and can be computed, along with its derivatives up to order ... at every given point), there is no such phenomenon as “local quadratic convergence”, the Newton method (which uses the second derivatives) has no advantages as compared to the methods which use only the first order derivatives, etc.;
- The very idea to reduce “black-box-represented” constrained convex problems to unconstrained ones – from the complexity viewpoint, the unconstrained problems are not easier than the constrained ones...

4.2.3 But...

Luckily, the pessimistic analysis of the classical interior penalty scheme is not the “final truth”. It turned out that what prevents this scheme to yield a polynomial time method is not the structure of the scheme, but the huge amount of freedom it allows for its elements (too much freedom is another word for anarchy...). After some order is added, the scheme becomes a polynomial time one! Specifically, it was understood that

1. There is a (completely non-traditional) class of “good” (self-concordant⁴⁾) barriers. Every barrier F of this type is associated with a “self-concordance parameter” $\theta(F)$, which is a real ≥ 1 ;
2. Whenever a barrier F underlying the interior penalty scheme is self-concordant, one can specify the notion of “closeness to the central path” and the policy for updating the penalty parameter in such a way that a *single* Newton step

$$x_i \mapsto x_{i+1} = x_i - [\nabla^2 F_{t_{i+1}}(x_i)]^{-1} \nabla F_{t_{i+1}}(x_i) \quad (4.2.1)$$

suffices to update a “close to $x_*(t_i)$ ” iterate x_i into a new iterate x_{i+1} which is close, in the same sense, to $x_*(t_{i+1})$. All “close to the central path” points belong to $\text{int } \mathcal{X}$, so that the scheme keeps all the iterates strictly feasible.

⁴⁾ We do not intend to explain here what is a “self-concordant barrier”; for our purposes it suffices to say that this is a three times continuously differentiable convex barrier F satisfying a pair of specific differential inequalities linking the first, the second and the third directional derivatives of F .

3. The penalty updating policy mentioned in the previous item is quite simple:

$$t_i \mapsto t_{i+1} = \left(1 + \frac{0.1}{\sqrt{\theta(F)}}\right) t_i;$$

in particular, it does not “slow down” as t_i grows and ensures linear, with the ratio $\left(1 + \frac{0.1}{\sqrt{\theta(F)}}\right)$, growth of the penalty. This is vitally important due to the following fact:

4. The inaccuracy of a point x , which is close to some point $x_*(t)$ of the central path, as an approximate solution to (C) is inverse proportional to t :

$$c^T x - \min_{y \in \mathcal{X}} c^T y \leq \frac{2\theta(F)}{t}.$$

It follows that

(!) After we have managed once to get close to the central path – have built a point x_0 which is close to a point $x(t_0)$, $t_0 > 0$, on the path, every $O(\sqrt{\theta(F)})$ steps of the scheme improve the quality of approximate solutions generated by the scheme by an absolute constant factor. In particular, it takes no more than

$$O(1)\sqrt{\theta(F)} \ln \left(2 + \frac{\theta(F)}{t_0 \epsilon}\right)$$

steps to generate a strictly feasible ϵ -solution to (C).

Note that with our simple penalty updating policy all needed to perform a step of the interior penalty scheme is to compute the gradient and the Hessian of the underlying barrier at a single point and to invert the resulting Hessian.

Items 3, 4 say that essentially all we need to derive from the just listed general results a polynomial time method for a generic convex optimization problem is to be able to equip every instance of the problem with a “good” barrier in such a way that both the parameter of self-concordance of the barrier $\theta(F)$ and the arithmetic cost at which we can compute the gradient and the Hessian of this barrier at a given point are polynomial in the size of the instance⁵⁾. And it turns out that we can meet the latter requirement for all interesting “well-structured” generic convex programs, in particular, for Linear, Conic Quadratic, and Semidefinite Programming. Moreover, “the heroes” of our course – LP, CQP and SDP – are especially nice application fields of the general theory of interior point polynomial time methods; in these particular applications, the theory can be simplified, on one hand, and strengthened, on another.

4.3 Interior point methods for LP, CQP, and SDP: building blocks

We are about to explain what the interior point methods for LP, CQP, SDP look like.

⁵⁾ Another requirement is to be able once get close to a point $x_*(t_0)$ on the central path with a not “disastrously small” value of t_0 – we should initialize somehow our path-following method! It turns out that such an initialization is a minor problem – it can be carried out via the same path-following technique, provided we are given in advance a strictly feasible solution to our problem.

4.3.1 Canonical cones and canonical barriers

We will be interested in a generic conic problem

$$\min_x \left\{ c^T x : Ax - B \in \mathbf{K} \right\} \quad (\text{CP})$$

associated with a cone \mathbf{K} given as a direct product of m “basic” cones, each of them being either a second-order, or a semidefinite cone:

$$\mathbf{K} = \mathbf{S}_+^{k_1} \times \dots \times \mathbf{S}_+^{k_p} \times \mathbf{L}^{k_{p+1}} \times \dots \times \mathbf{L}^{k_m} \subset E = \mathbf{S}^{k_1} \times \dots \times \mathbf{S}^{k_p} \times \mathbf{R}^{k_{p+1}} \times \dots \times \mathbf{R}^{k_m}. \quad (\text{Cone})$$

Of course, the generic problem in question covers LP (no Lorentz factors, all semidefinite factors are of dimension 1), CQP (no semidefinite factors) and SDP (no Lorentz factors).

Now, we shall equip the semidefinite and the Lorentz cones with “canonical barriers”:

- The canonical barrier for a semidefinite cone \mathbf{S}_+^n is

$$S_k(X) = -\ln \text{Det}(X) : \text{int } \mathbf{S}_+^k \rightarrow \mathbf{R};$$

the parameter of this barrier, by definition, is $\theta(S_k) = k$ ⁶⁾.

- the canonical barrier for a Lorentz cone $\mathbf{L}^k = \{x \in \mathbf{R}^k \mid x_k \geq \sqrt{x_1^2 + \dots + x_{k-1}^2}\}$ is

$$L_k(x) = -\ln(x_k^2 - x_1^2 - \dots - x_{k-1}^2) = -\ln(x^T J_k x), \quad J_k = \begin{pmatrix} -I_{k-1} & \\ & 1 \end{pmatrix};$$

the parameter of this barrier is $\theta(L_k) = 2$.

- The canonical barrier K for the cone \mathbf{K} given by (Cone), by definition, is the direct sum of the canonical barriers of the factors:

$$K(X) = S_{k_1}(X_1) + \dots + S_{k_p}(X_p) + L_{k_{p+1}}(X_{p+1}) + \dots + L_{k_m}(X_m), \quad X_i \in \begin{cases} \text{int } \mathbf{S}_+^{k_i}, & i \leq p \\ \text{int } \mathbf{L}^{k_i}, & p < i \leq m \end{cases};$$

from now on, we use upper case Latin letters, like X, Y, Z , to denote elements of the space E ; for such an element X , X_i denotes the projection of X onto i -th factor in the direct product representation of E as shown in (Cone).

The parameter of the barrier K , again by definition, is the sum of parameters of the basic barriers involved:

$$\theta(K) = \theta(S_{k_1}) + \dots + \theta(S_{k_p}) + \theta(L_{k_{p+1}}) + \dots + \theta(L_{k_m}) = \sum_{i=1}^p k_i + 2(m-p).$$

Recall that all direct factors in the direct product representation (Cone) of our “universe” E are Euclidean spaces; the matrix factors \mathbf{S}^{k_i} are endowed with the Frobenius inner product

$$\langle X_i, Y_i \rangle_{\mathbf{S}^{k_i}} = \text{Tr}(X_i Y_i),$$

while the “arithmetic factors” \mathbf{R}^{k_i} are endowed with the usual inner product

$$\langle X_i, Y_i \rangle_{\mathbf{R}^{k_i}} = X_i^T Y_i;$$

⁶⁾ The barrier S^k , same as the canonical barrier L_k for the Lorentz cone \mathbf{L}^k , indeed are self-concordant (whatever it means), and the parameters they are assigned here by definition are exactly their parameters of self-concordance.

E itself will be regarded as a Euclidean space endowed with the direct sum of inner products on the factors:

$$\langle X, Y \rangle_E = \sum_{i=1}^p \text{Tr}(X_i Y_i) + \sum_{i=p+1}^m X_i^T Y_i.$$

It is clearly seen that our basic barriers, same as their direct sum K , indeed are barriers for the corresponding cones: they are C^∞ -smooth on the interiors of their domains, blow up to ∞ along every sequence of points from these interiors converging to a boundary point of the corresponding domain and are strongly convex. To verify the latter property, it makes sense to compute explicitly the first and the second directional derivatives of these barriers (we need the corresponding formulae in any case); to simplify notation, we write down the derivatives of the basic functions S_k, L_k at a point x from their domain along a direction h (you should remember that in the case of S_k both the point and the direction, in spite of their lower-case denotation, are $k \times k$ symmetric matrices):

	$DS_k(x)[h] \equiv \frac{d}{dt} \Big _{t=0} S_k(x + th) = -\text{Tr}(x^{-1}h) = -\langle x^{-1}, h \rangle_{\mathbf{S}^k},$	(4.3.1)
	i.e.	
	$\nabla S_k(x) = -x^{-1};$	
	$D^2 S_k(x)[h, h] \equiv \frac{d^2}{dt^2} \Big _{t=0} S_k(x + th) = \text{Tr}(x^{-1}hx^{-1}h) = \langle x^{-1}hx^{-1}, h \rangle_{\mathbf{S}^k},$	
	i.e.	
	$[\nabla^2 S_k(x)]h = x^{-1}hx^{-1};$	
	$DL_k(x)[h] \equiv \frac{d}{dt} \Big _{t=0} L_k(x + th) = -2\frac{h^T J_k x}{x^T J_k x},$	
	i.e.	
	$\nabla L_k(x) = -\frac{2}{x^T J_k x} J_k x;$	
	$D^2 L_k(x)[h, h] \equiv \frac{d^2}{dt^2} \Big _{t=0} L_k(x + th) = 4\frac{[h^T J_k x]^2}{[x^T J_k x]^2} - 2\frac{h^T J_k h}{x^T J_k x},$	
	i.e.	
	$\nabla^2 L_k(x) = \frac{4}{[x^T J_k x]^2} J_k x x^T J_k - \frac{2}{x^T J_k x} J_k.$	

From the expression for $D^2 S_k(x)[h, h]$ we see that

$$D^2 S_k(x)[h, h] = \text{Tr}(x^{-1}hx^{-1}h) = \text{Tr}([x^{-1/2}hx^{-1/2}]^2),$$

so that $D^2 S_k(x)[h, h]$ is positive whenever $h \neq 0$. It is not difficult to prove that the same is true for $D^2 L_k(x)[h, h]$. Thus, the canonical barriers for semidefinite and Lorentz cones are strongly convex, and so is their direct sum $K(\cdot)$.

It makes sense to illustrate relatively general concepts and results to follow by how they look in a particular case when \mathbf{K} is the semidefinite cone \mathbf{S}_+^k ; we shall refer to this situation as to the “SDP case”. The essence of the matter in our general case is exactly the same as in this particular one, but “straightforward computations” which are easy in the SDP case become nearly impossible in the general case; and we have no possibility to explain here how it is possible (it is!) to get the desired results with minimum amount of computations.

Due to the role played by the SDP case in our exposition, we use for this case special notation, along with the just introduced “general” one. Specifically, we denote the standard – the Frobenius – inner product on $E = \mathbf{S}^k$ as $\langle \cdot, \cdot \rangle_F$, although feel free, if necessary, to use our “general” notation $\langle \cdot, \cdot \rangle_E$ as well; the associated norm is denoted by $\|\cdot\|_2$, so that $\|X\|_2 = \sqrt{\text{Tr}(X^2)}$, X being a symmetric matrix.

4.3.2 Elementary properties of canonical barriers

Let us establish a number of simple and useful properties of canonical barriers.

Proposition 4.3.1 *A canonical barrier, let it be denoted F (F can be either S_k , or L_k , or the direct sum K of several copies of these “elementary” barriers), possesses the following properties:*

(i) *F is logarithmically homogeneous, the parameter of logarithmic homogeneity being $-\theta(F)$, i.e., the following identity holds:*

$$t > 0, x \in \text{Dom } F \Rightarrow F(tx) = F(x) - \theta(F) \ln t.$$

- In the SDP case, i.e., when $F = S_k = -\ln \text{Det}(x)$ and x is $k \times k$ positive definite matrix, (i) claims that

$$-\ln \text{Det}(tx) = -\ln \text{Det}(x) - k \ln t,$$

which of course is true.

(ii) *Consequently, the following two equalities hold identically in $x \in \text{Dom } F$:*

- (a) $\langle \nabla F(x), x \rangle = -\theta(F);$
- (b) $[\nabla^2 F(x)]x = -\nabla F(x).$

- In the SDP case, $\nabla F(x) = \nabla S_k(x) = -x^{-1}$ and $[\nabla^2 F(x)]h = \nabla^2 S_k(x)h = x^{-1}hx^{-1}$ (see (4.3.1)). Here (a) becomes the identity $\langle x^{-1}, x \rangle_F \equiv \text{Tr}(x^{-1}x) = k$, and (b) kindly informs us that $x^{-1}xx^{-1} = x^{-1}$.

(iii) *Consequently, k -th differential $D^k F(x)$ of F , $k \geq 1$, is homogeneous, of degree $-k$, in $x \in \text{Dom } F$:*

$\forall (x \in \text{Dom } F, t > 0, h_1, \dots, h_k) :$

$$D^k F(tx)[h_1, \dots, h_k] \equiv \frac{\partial^k F(tx + s_1 h_1 + \dots + s_k h_k)}{\partial s_1 \partial s_2 \dots \partial s_k} \Big|_{s_1 = \dots = s_k = 0} = t^{-k} D^k F(x)[h_1, \dots, h_k]. \quad (4.3.2)$$

Proof. (i): it is immediately seen that S_k and L_k are logarithmically homogeneous with parameters of logarithmic homogeneity $-\theta(S_k)$, $-\theta(L_k)$, respectively; and of course the property of logarithmic homogeneity is stable with respect to taking direct sums of functions: if $\text{Dom } \Phi(u)$ and $\text{Dom } \Psi(v)$ are closed w.r.t. the operation of multiplying a vector by a positive scalar, and both Φ and Ψ are logarithmically homogeneous with parameters α , β , respectively, then the function $\Phi(u) + \Psi(v)$ is logarithmically homogeneous with the parameter $\alpha + \beta$.

(ii): To get (ii.a), it suffices to differentiate the identity

$$F(tx) = F(x) - \theta(F) \ln t$$

in t at $t = 1$:

$$F(tx) = F(x) - \theta(F) \ln t \Rightarrow \langle \nabla F(tx), x \rangle = \frac{d}{dt} F(tx) = -\theta(F)t^{-2},$$

and it remains to set $t = 1$ in the concluding identity.

Similarly, to get (ii.b), it suffices to differentiate the identity

$$\langle \nabla F(x + th), x + th \rangle = -\theta(F)$$

(which is just (ii.a)) in t at $t = 0$, thus arriving at

$$\langle [\nabla^2 F(x)]h, x \rangle + \langle \nabla F(x), h \rangle = 0;$$

since $\langle [\nabla^2 F(x)]h, x \rangle = \langle [\nabla^2 F(x)]x, h \rangle$ (symmetry of partial derivatives!) and since the resulting equality

$$\langle [\nabla^2 F(x)]x, h \rangle + \langle \nabla F(x), h \rangle = 0$$

holds true identically in h , we come to $[\nabla^2 F(x)]x = -\nabla F(x)$.

(iii): Differentiating k times the identity

$$F(tx) = F(x) - \theta \ln t$$

in x , we get

$$t^k D^k F(tx)[h_1, \dots, h_k] = D^k F(x)[h_1, \dots, h_k]. \quad \blacksquare$$

An especially nice specific feature of the barriers S_k , L_k and K is their *self-duality*:

Proposition 4.3.2 *A canonical barrier, let it be denoted F (F can be either S_k , or L_k , or the direct sum K of several copies of these “elementary” barriers), possesses the following property: for every $x \in \text{Dom } F$, $-\nabla F(x)$ belongs to $\text{Dom } F$ as well, and the mapping $x \mapsto -\nabla F(x) : \text{Dom } F \rightarrow \text{Dom } F$ is self-inverse:*

$$-\nabla F(-\nabla F(x)) = x \quad \forall x \in \text{Dom } F. \quad (4.3.3)$$

Besides this, the mapping $x \mapsto -\nabla F(x)$ is homogeneous of degree -1:

$$t > 0, x \in \text{int dom } F \Rightarrow -\nabla F(tx) = -t^{-1} \nabla F(x). \quad (4.3.4)$$

- In the SDP case, i.e., when $F = S_k$ and x is $k \times k$ semidefinite matrix, $\nabla F(x) = \nabla S_k(x) = -x^{-1}$, see (4.3.1), so that the above statements merely say that the mapping $x \mapsto x^{-1}$ is a self-inverse one-to-one mapping of the interior of the semidefinite cone onto itself, and that $-(tx)^{-1} = -t^{-1}x^{-1}$, both claims being trivially true.

4.4 Primal-dual pair of problems and primal-dual central path

4.4.1 The problem(s)

It makes sense to consider simultaneously the “problem of interest” (CP) and its conic dual; since \mathbf{K} is a direct product of self-dual cones, this dual is a conic problem *on the same cone \mathbf{K}* . As we remember from Lecture 1, the primal-dual pair associated with (CP) is

$$\begin{aligned} \min_x \{c^T x : \mathcal{A}x - B \in \mathbf{K}\} \quad & (\text{CP}) \\ \max_S \{\langle B, S \rangle_E : \mathcal{A}^*S = c, S \in \mathbf{K}\} \quad & (\text{CD}) \end{aligned}$$

Assuming that the linear mapping $x \mapsto \mathcal{A}x$ is an embedding (i.e., that $\text{Ker } \mathcal{A} = \{0\}$ – this is Assumption A from Lecture 1), we can write down our primal-dual pair in a symmetric geometric form (Lecture 1, Section 1.4.4):

$$\begin{aligned} \min_X \{\langle C, X \rangle_E : X \in (\mathcal{L} - B) \cap \mathbf{K}\} \quad & (\text{P}) \\ \max_S \{\langle B, S \rangle_E : S \in (\mathcal{L}^\perp + C) \cap \mathbf{K}\} \quad & (\text{D}) \end{aligned}$$

where \mathcal{L} is a linear subspace in E (the image space of the linear mapping $x \mapsto Ax$), \mathcal{L}^\perp is the orthogonal complement to \mathcal{L} in E , and $C \in E$ satisfies $A^*C = c$, i.e., $\langle C, Ax \rangle_E \equiv c^T x$.

To simplify things, from now on we assume that both problems (CP) and (CD) are strictly feasible. In terms of (P) and (D) this assumption means that both the primal feasible plane $\mathcal{L} - B$ and the dual feasible plane $\mathcal{L}^\perp + C$ intersect the interior of the cone \mathbf{K} .

Remark 4.4.1 By the Conic Duality Theorem (Lecture 1), both (CP) and (D) are solvable with equal optimal values:

$$\text{Opt(CP)} = \text{Opt(D)}$$

(recall that we have assumed strict primal-dual feasibility). Since (P) is equivalent to (CP), (P) is solvable as well, and the optimal value of (P) differs from the one of (P) by $\langle C, B \rangle_E$ ⁷⁾. It follows that the optimal values of (P) and (D) are linked by the relation

$$\text{Opt(P)} - \text{Opt(D)} + \langle C, B \rangle_E = 0. \quad (4.4.1)$$

4.4.2 The central path(s)

The canonical barrier K of \mathbf{K} induces a barrier for the feasible set $\mathcal{X} = \{x \mid Ax - B \in \mathbf{K}\}$ of the problem (CP) written down in the form of (C), i.e., as

$$\min_x \{c^T x : x \in \mathcal{X}\};$$

this barrier is

$$\widehat{K}(x) = K(Ax - B) : \text{int } X \rightarrow \mathbf{R} \quad (4.4.2)$$

and is indeed a barrier. Now we can apply the interior penalty scheme to trace the central path $x_*(t)$ associated with the resulting barrier; with some effort it can be derived from the primal-dual strict feasibility that this central path is well-defined (i.e., that the minimizer of

$$\widehat{K}_t(x) = tc^T x + \widehat{K}(x)$$

on $\text{int } X$ exists for every $t \geq 0$ and is unique⁸⁾. What is important for us for the moment, is the central path itself, not how to trace it. Moreover, it is highly instructive to pass from the central path $x_*(t)$ in the space of design variables to its image

$$X_*(t) = Ax_*(t) - B$$

in E . The resulting curve has a name – it is called *the primal central path of the primal-dual pair (P), (D)*; by its origin, it is a curve comprised of strictly feasible solutions of (P) (since it is the same – to say that x belongs to the (interior of) the set \mathcal{X} and to say that $X = Ax - B$ is a (strictly) feasible solution of (P)). A simple and very useful observation is that the primal central path can be defined solely in terms of (P), (D) and thus is a “geometric entity” – it is independent of a particular parameterization of the primal feasible plane $\mathcal{L} - B$ by the design vector x :

⁷⁾ Indeed, the values of the respective objectives $c^T x$ and $\langle C, Ax - B \rangle_E$ at the corresponding to each other feasible solutions x of (CP) and $X = Ax - B$ of (P) differ from each other by exactly $\langle C, B \rangle_E$:

$$c^T x - \langle C, X \rangle_E = c^T x - \langle C, Ax - B \rangle_E = \underbrace{c^T x - \langle A^*C, x \rangle_E}_{=0 \text{ due to } A^*C=c} + \langle C, B \rangle_E.$$

⁸⁾ In Section 4.2.1, there was no problem with the existence of the central path, since there \mathcal{X} was assumed to be bounded; in our now context, \mathcal{X} not necessarily is bounded.

(*) A point $X_*(t)$ of the primal central path is the minimizer of the aggregate

$$P_t(X) = t\langle C, X \rangle_E + K(X)$$

on the set $(\mathcal{L} - B) \cap \text{int } \mathbf{K}$ of strictly feasible solutions of (P).

This observation is just a tautology: $x_*(t)$ is the minimizer on $\text{int } \mathcal{X}$ of the aggregate

$$\widehat{K}_t(x) \equiv tc^T x + \widehat{K}(x) = t\langle C, Ax \rangle_E + K(Ax - B) = P_t(Ax - B) + t\langle C, B \rangle_E;$$

we see that the function $\widehat{P}_t(x) = P_t(Ax - B)$ of $x \in \text{int } \mathcal{X}$ differs from the function $\widehat{K}_t(x)$ by a constant (depending on t) and has therefore the same minimizer $x_*(t)$ as the function $\widehat{K}_t(x)$. Now, when x runs through $\text{int } \mathcal{X}$, the point $X = Ax - B$ runs exactly through the set of strictly feasible solutions of (P), so that the minimizer X_* of P_t on the latter set and the minimizer $x_*(t)$ of the function $\widehat{P}_t(x) = P_t(Ax - B)$ on $\text{int } \mathcal{X}$ are linked by the relation $X_* = Ax_*(t) - B$.

The “analytic translation” of the above observation is as follows:

(*)' A point $X_*(t)$ of the primal central path is exactly the strictly feasible solution X to (P) such that the vector $tC + \nabla K(X) \in E$ is orthogonal to \mathcal{L} (i.e., belongs to \mathcal{L}^\perp).

Indeed, we know that $X_*(t)$ is the unique minimizer of the smooth convex function $P_t(X) = t\langle C, X \rangle_E + K(X)$ on the intersection of the primal feasible plane $\mathcal{L} - B$ and the interior of the cone \mathbf{K} ; a necessary and sufficient condition for a point X of this intersection to minimize P_t over the intersection is that ∇P_t must be orthogonal to \mathcal{L} .

- In the SDP case, a point $X_*(t)$, $t > 0$, of the primal central path is uniquely defined by the following two requirements: (1) $X_*(t) \succ 0$ should be feasible for (P), and (2) the $k \times k$ matrix

$$tC - X_*^{-1}(t) = tC + \nabla S_k(X_*(t))$$

(see (4.3.1)) should belong to \mathcal{L}^\perp , i.e., should be orthogonal, w.r.t. the Frobenius inner product, to every matrix of the form Ax .

The dual problem (D) is in no sense “worse” than the primal problem (P) and thus also possesses the central path, now called the *dual central path* $S_*(t)$, $t \geq 0$, of the primal-dual pair (P), (D). Similarly to (*), (*'), the dual central path can be characterized as follows:

(**) A point $S_*(t)$, $t \geq 0$, of the dual central path is the unique minimizer of the aggregate

$$D_t(S) = -t\langle B, S \rangle_E + K(S)$$

on the set of strictly feasible solutions of (D)⁹⁾. $S_*(t)$ is exactly the strictly feasible solution S to (D) such that the vector $-tB + \nabla F(S)$ is orthogonal to \mathcal{L}^\perp (i.e., belongs to \mathcal{L}).

⁹⁾ Note the slight asymmetry between the definitions of the primal aggregate P_t and the dual aggregate D_t : in the former, the linear term is $t\langle C, X \rangle_E$, while in the latter it is $-t\langle B, S \rangle_E$. This asymmetry is in complete accordance with the fact that we write (P) as a minimization, and (D) – as a maximization problem; to write (D) in exactly the same form as (P), we were supposed to replace B with $-B$, thus getting the formula for D_t completely similar to the one for P_t .

- In the SDP case, a point $S_*(t)$, $t > 0$, of the dual central path is uniquely defined by the following two requirements: (1) $S_*(t) \succ 0$ should be feasible for (D), and (2) the $k \times k$ matrix

$$-tB - S_*^{-1}(t) = -tB + \nabla S_k(S_*(t))$$

(see (4.3.1)) should belong to \mathcal{L} , i.e., should be representable in the form $\mathcal{A}x$ for some x .

From Proposition 4.3.2 we can derive a wonderful connection between the primal and the dual central paths:

Theorem 4.4.1 *For $t > 0$, the primal and the dual central paths $X_*(t)$, $S_*(t)$ of a (strictly feasible) primal-dual pair (P), (D) are linked by the relations*

$$\begin{aligned} S_*(t) &= -t^{-1}\nabla K(X_*(t)) \\ X_*(t) &= -t^{-1}\nabla K(S_*(t)) \end{aligned} \tag{4.4.3}$$

Proof. By (*'), the vector $tC + \nabla K(X_*(t))$ belongs to \mathcal{L}^\perp , so that the vector $S = -t^{-1}\nabla K(X_*(t))$ belongs to the dual feasible plane $\mathcal{L}^\perp + C$. On the other hand, by Proposition 4.4.3 the vector $-\nabla K(X_*(t))$ belongs to $\text{Dom } K$, i.e., to the interior of \mathbf{K} ; since \mathbf{K} is a cone and $t > 0$, the vector $S = -t^{-1}\nabla F(X_*(t))$ belongs to the interior of \mathbf{K} as well. Thus, S is a strictly feasible solution of (D). Now let us compute the gradient of the aggregate D_t at the point S :

$$\begin{aligned} \nabla D_t(S) &= -tB + \nabla K(-t^{-1}\nabla K(X_*(t))) \\ &= -tB + t\nabla K(-\nabla K(X_*(t))) \\ &\quad [\text{we have used (4.3.4)}] \\ &= -tB - tX_*(t) \\ &\quad [\text{we have used (4.3.3)}] \\ &= -t(B + X_*(t)) \\ &\in \mathcal{L} \\ &\quad [\text{since } X_*(t) \text{ is primal feasible}] \end{aligned}$$

Thus, S is strictly feasible for (D) and $\nabla D_t(S) \in \mathcal{L}$. But by (**') these properties characterize $S_*(t)$; thus, $S_*(t) = S \equiv -t^{-1}\nabla K(X_*(t))$. This relation, in view of Proposition 4.3.2, implies that $X_*(t) = -t^{-1}\nabla K(S_*(t))$. Another way to get the latter relation from the one $S_*(t) = -t^{-1}\nabla K(X_*(t))$ is just to refer to the primal-dual symmetry. ■

In fact, the connection between the primal and the dual central paths stated by Theorem 4.4.1 can be used to characterize both the paths:

Theorem 4.4.2 *Let (P), (D) be a strictly feasible primal-dual pair.*

For every $t > 0$, there exists a unique strictly feasible solution X of (P) such that $-t^{-1}\nabla K(X)$ is a feasible solution to (D), and this solution X is exactly $X_(t)$.*

Similarly, for every $t > 0$, there exists a unique strictly feasible solution S of (D) such that $-t^{-1}\nabla K(S)$ is a feasible solution of (P), and this solution S is exactly $S_(t)$.*

Proof. By primal-dual symmetry, it suffices to prove the first claim. We already know (Theorem 4.4.1) that $X = X_*(t)$ is a strictly feasible solution of (P) such that $-t^{-1}\nabla K(X)$ is feasible for (D); all we need to prove is that $X_*(t)$ is the only point with these properties, which is immediate: if X is a strictly feasible solution of (P) such that $-t^{-1}\nabla K(X)$ is dual feasible, then $-t^{-1}\nabla K(X) \in \mathcal{L}^\perp + C$, or, which is the same, $\nabla K(X) \in \mathcal{L}^\perp - tC$, or, which again is the same, $\nabla P_t(X) = tC + \nabla K(X) \in \mathcal{L}^\perp$. And we already know from (*) that the latter property, taken together with the strict primal feasibility, is characteristic for $X_*(t)$. ■

4.4.2.1 On the central path

As we have seen, the primal and the dual central paths are intrinsically linked one to another, and it makes sense to think of them as of a unique entity – the *primal-dual* central path of the primal-dual pair (P), (D). The primal-dual central path is just a curve $(X_*(t), S_*(t))$ in $E \times E$ such that the projection of the curve on the primal space is the primal central path, and the projection of it on the dual space is the dual central path.

To save words, from now on we refer to the primal-dual central path simply as to the *central path*.

The central path possesses a number of extremely nice properties; let us list some of them.

Characterization of the central path. By Theorem 4.4.2, the points $(X_*(t), S_*(t))$ of the central path possess the following properties:

(CentralPath):

1. [Primal feasibility] *The point $X_*(t)$ is strictly primal feasible.*
2. [Dual feasibility] *The point $S_*(t)$ is dual feasible.*
3. [“Augmented complementary slackness”] The points $X_*(t)$ and $S_*(t)$ are linked by the relation

$$S_*(t) = -t^{-1}\nabla K(X_*(t)) \quad [\Leftrightarrow X_*(t) = -t^{-1}\nabla K(S_*(t))].$$

- In the SDP case, $\nabla K(U) = \nabla S_k(U) = -U^{-1}$ (see (4.3.1)), and the augmented complementary slackness relation takes the nice form

$$X_*(t)S_*(t) = t^{-1}I, \tag{4.4.4}$$

where I , as usual, is the unit matrix.

In fact, the indicated properties *fully characterize* the central path: whenever two points X, S possess the properties 1) - 3) with respect to some $t > 0$, X is nothing but $X_*(t)$, and S is nothing but $S_*(t)$ (this again is said by Theorem 4.4.2).

Duality gap along the central path. Recall that for an arbitrary primal-dual feasible pair (X, S) of the (strictly feasible!) primal-dual pair of problems (P), (D), the duality gap

$$\text{DualityGap}(X, S) \equiv [\langle C, X \rangle_E - \text{Opt}(P)] + [\text{Opt}(D) - \langle B, S \rangle_E] = \langle C, X \rangle_E - \langle B, S \rangle_E + \langle C, B \rangle_E$$

(see (4.4.1)) which measures the “total inaccuracy” of X, S as approximate solutions of the respective problems, can be written down equivalently as $\langle S, X \rangle_E$ (see statement (!) in Section 1.4.5). Now, what is the duality gap along the central path? The answer is immediate:

$$\begin{aligned} \text{DualityGap}(X_*(t), S_*(t)) &= \langle S_*(t), X_*(t) \rangle_E \\ &= \langle -t^{-1}\nabla K(X_*(t)), X_*(t) \rangle_E \\ &\quad [\text{see (4.4.3)}] \\ &= t^{-1}\theta(K) \\ &\quad [\text{see Proposition 4.3.1.(ii)}] \end{aligned}$$

We have arrived at a wonderful result¹⁰⁾:

Proposition 4.4.1 *Under assumption of primal-dual strict feasibility, the duality gap along the central path is inverse proportional to the penalty parameter, the proportionality coefficient being the parameter of the canonical barrier K :*

$$\text{DualityGap}(X_*(t), S_*(t)) = \frac{\theta(K)}{t}.$$

In particular, both $X_*(t)$ and $S_*(t)$ are strictly feasible $\left(\frac{\theta(K)}{t}\right)$ -approximate solutions to their respective problems:

$$\begin{aligned} \langle C, X_*(t) \rangle_E - \text{Opt}(P) &\leq \frac{\theta(K)}{t}, \\ \text{Opt}(D) - \langle B, S_*(t) \rangle_E &\leq \frac{\theta(K)}{t}. \end{aligned}$$

- In the SDP case, $\mathbf{K} = \mathbf{S}_+^k$ and $\theta(K) = \theta(S_k) = k$.

We see that

All we need in order to get “quickly” good primal and dual approximate solutions, is to trace fast the central path; if we were interested to solve only one of the problems (P) , (D) , it would be sufficient to trace fast the associated – primal or dual – component of this path. The quality guarantees we get in such a process depend – in a completely universal fashion! – solely on the value t of the penalty parameter we have managed to achieve and on the value of the parameter of the canonical barrier K and are completely independent of other elements of the data.

4.4.2.2 Near the central path

The conclusion we have just made is a bit too optimistic: well, our life when moving along the central path would be just fine (at the very least, we would know how good are the solutions we already have), but how could we move exactly along the path? Among the relations (CentralPath.1-3) defining the path the first two are “simple” – just linear, but the third is in fact a system of *nonlinear* equations, and we have no hope to satisfy these equations exactly. Thus, we arrive at the crucial question which, a bit informally, sounds as follows:

How close (and in what sense close) should we be to the path in order for our life to be essentially as nice as if we were exactly on the path?

There are several ways to answer this question; we will present the simplest one.

A distance to the central path. Our canonical barrier $K(\cdot)$ is a strongly convex smooth function on $\text{int } \mathbf{K}$; in particular, its Hessian matrix $\nabla^2 K(Y)$, taken at a point $Y \in \text{int } \mathbf{K}$, is positive definite. We can use the inverse of this matrix to measure the distances between points of E , thus arriving at the norm

$$\|H\|_Y = \sqrt{\langle [\nabla^2 K(Y)]^{-1} H, H \rangle_E}.$$

It turns out that

¹⁰⁾ Which, among other, much more important consequences, explains the name “augmented complementary slackness” of the property 1^{0.3}): at the primal-dual pair of *optimal* solutions X^*, S^* the duality gap should be zero: $\langle S^*, X^* \rangle_E = 0$. Property 1^{0.3}, as we just have seen, implies that the duality gap at a primal-dual pair $(X_*(t), S_*(t))$ from the central path, although nonzero, is “controllable” – $\frac{\theta(K)}{t}$ – and becomes small as t grows.

A good measure of proximity of a strictly feasible primal-dual pair $Z = (X, S)$ to a point $Z_*(t) = (X_*(t), S_*(t))$ from the primal-dual central path is the quantity

$$\text{dist}(Z, Z_*(t)) \equiv \|tS + \nabla K(X)\|_X \equiv \sqrt{\langle [\nabla^2 K(X)]^{-1}(tS + \nabla K(X)), tS + \nabla K(X) \rangle_E}$$

Although written in a non-symmetric w.r.t. X, S form, this quantity is in fact symmetric in X, S : it turns out that

$$\|tS + \nabla K(X)\|_X = \|tX + \nabla K(S)\|_S \quad (4.4.5)$$

for all $t > 0$ and $S, X \in \text{int } \mathbf{K}$.

Observe that $\text{dist}(Z, Z_*(t)) \geq 0$, and $\text{dist}(Z, Z_*(t)) = 0$ if and only if $S = -t^{-1}\nabla K(X)$, which, for a strictly primal-dual feasible pair $Z = (X, S)$, means that $Z = Z_*(t)$ (see the characterization of the primal-dual central path); thus, $\text{dist}(Z, Z_*(t))$ indeed can be viewed as a kind of distance from Z to $Z_*(t)$.

In the SDP case X, S are $k \times k$ symmetric matrices, and

$$\begin{aligned} \text{dist}^2(Z, Z_*(t)) &= \|tS + \nabla S_k(X)\|_X^2 = \langle [\nabla^2 S_k(X)]^{-1}(tS + \nabla S_k(X)), tS + \nabla S_k(X) \rangle_F \\ &= \text{Tr}(X(tS - X^{-1})X(tS - X^{-1})) \quad [\text{see (4.3.1)}] \\ &= \text{Tr}([tX^{1/2}SX^{1/2} - I]^2), \end{aligned}$$

so that

$$\text{dist}^2(Z, Z_*(t)) = \text{Tr}(X(tS - X^{-1})X(tS - X^{-1})) = \|tX^{1/2}SX^{1/2} - I\|_2^2. \quad (4.4.6)$$

Besides this,

$$\begin{aligned} \|tX^{1/2}SX^{1/2} - I\|_2^2 &= \text{Tr}([tX^{1/2}SX^{1/2} - I]^2) \\ &= \text{Tr}(t^2X^{1/2}SX^{1/2}X^{1/2}SX^{1/2} - 2tX^{1/2}SX^{1/2} + I) \\ &= \text{Tr}(t^2X^{1/2}SXSX^{1/2}) - 2t\text{Tr}(X^{1/2}SX^{1/2}) + \text{Tr}(I) \\ &= \text{Tr}(t^2XSXS - 2tXS + I) \\ &= \text{Tr}(t^2SXSX - 2tSX + I) \\ &= \text{Tr}(t^2S^{1/2}XS^{1/2}S^{1/2}XS^{1/2} - 2tS^{1/2}XS^{1/2} + I) \\ &= \text{Tr}([tS^{1/2}XS^{1/2} - I]^2), \end{aligned}$$

i.e., (4.4.5) indeed is true.

In a moderate $\text{dist}(\cdot, Z_*(\cdot))$ -neighbourhood of the central path. It turns out that in such a neighbourhood all is essentially as fine as at the central path itself:

A. Whenever $Z = (X, S)$ is a pair of primal-dual strictly feasible solutions to (P), (D) such that

$$\text{dist}(Z, Z_*(t)) \leq 1, \quad (\text{Close})$$

Z is “essentially as good as $Z_*(t)$ ”, namely, the duality gap at (X, S) is essentially as small as at the point $Z_*(t)$:

$$\text{DualityGap}(X, S) = \langle S, X \rangle_E \leq 2\text{DualityGap}(Z_*(t)) = \frac{2\theta(K)}{t}. \quad (4.4.7)$$

Let us check **A** in the SDP case. Let (t, X, S) satisfy the premise of **A**. The duality gap at the pair (X, S) of strictly primal-dual feasible solutions is

$$\text{DualityGap}(X, S) = \langle X, S \rangle_F = \text{Tr}(XS),$$

while by (4.4.6) the relation $\text{dist}((S, X), Z_*(t)) \leq 1$ means that

$$\|tX^{1/2}SX^{1/2} - I\|_2 \leq 1,$$

whence

$$\|X^{1/2}SX^{1/2} - t^{-1}I\|_2 \leq \frac{1}{t}.$$

Denoting by δ the vector of eigenvalues of the symmetric matrix $X^{1/2}SX^{1/2}$, we conclude

that $\sum_{i=1}^k (\delta_i - t^{-1})^2 \leq t^{-2}$, whence

$$\begin{aligned} \text{DualityGap}(X, S) &= \text{Tr}(XS) = \text{Tr}(X^{1/2}SX^{1/2}) = \sum_{i=1}^k \delta_i \\ &\leq kt^{-1} + \sum_{i=1}^k |\delta_i - t^{-1}| \leq kt^{-1} + \sqrt{k} \sqrt{\sum_{i=1}^k (\delta_i - t^{-1})^2} \\ &\leq kt^{-1} + \sqrt{kt^{-1}}, \end{aligned}$$

and (4.4.7) follows.

It follows from **A** that

For our purposes, it is essentially the same – to move along the primal-dual central path, or to trace this path, staying in its “time-space” neighbourhood

$$\mathcal{N}_\kappa = \{(t, X, S) \mid X \in \mathcal{L} - B, S \in \mathcal{L}^\perp + C, t > 0, \text{dist}((X, S), (X_*(t), S_*(t))) \leq \kappa\} \quad (4.4.8)$$

with certain $\kappa \leq 1$.

Most of the interior point methods for LP, CQP, and SDP, including those most powerful in practice, solve the primal-dual pair (P), (D) by tracing the central path¹¹⁾, although not all of them keep the iterates in $\mathcal{N}_{O(1)}$; some of the methods work in much wider neighbourhoods of the central path, in order to avoid slowing down when passing “highly curved” segments of the path. At the level of ideas, these “long step path following methods” essentially do not differ from the “short step” ones – those keeping the iterates in $\mathcal{N}_{O(1)}$; this is why in the analysis part of our forthcoming presentation we restrict ourselves with the short-step methods. It should be added that as far as the theoretical efficiency estimates are concerned, the short-step methods yield the best known so far complexity bounds for LP, CQP and SDP, and are essentially better than the long-step methods (although in practice the long-step methods usually outperform their short-step counterparts).

4.5 Tracing the central path

4.5.1 The path-following scheme

Assume we are solving a strictly feasible primal-dual pair of problems (P), (D) and intend to trace the associated central path. Essentially all we need is a mechanism for updating a current

¹¹⁾ There exist also *potential reduction* interior point methods which do not take explicit care of tracing the central path; an example is the very first IP method for LP – the method of Karmarkar. The potential reduction IP methods are beyond the scope of our course, which is not a big loss for a practically oriented reader, since, as a practical tool, these methods are thought of to be obsolete.

iterate $(\bar{t}, \bar{X}, \bar{S})$ such that $\bar{t} > 0$, \bar{X} is strictly primal feasible, \bar{S} is strictly dual feasible, and (\bar{X}, \bar{S}) is “good”, in certain precise sense, approximation of the point $Z_*(\bar{t}) = (X_*(\bar{t}), S_*(\bar{t}))$ on the central path, into a new iterate (t_+, X_+, S_+) with similar properties and a larger value $t_+ > \bar{t}$ of the penalty parameter. Given such an updating and iterating it, we indeed shall trace the central path, with all the benefits (see above) coming from the latter fact¹²⁾ How could we construct the required updating? Recalling the description of the central path, we see that our question is:

Given a triple $(\bar{t}, \bar{X}, \bar{S})$ which satisfies the relations

$$\begin{aligned} X &\in \mathcal{L} - B, \\ S &\in \mathcal{L}^\perp + C \end{aligned} \quad (4.5.1)$$

(which is in fact a system of linear equations) and approximately satisfies the system of nonlinear equations

$$G_t(X, S) \equiv S + t^{-1} \nabla K(X) = 0, \quad (4.5.2)$$

update it into a new triple (t_+, X_+, S_+) with the same properties and $t_+ > \bar{t}$.

Since the left hand side $G(\cdot)$ in our system of nonlinear equations is smooth around $(\bar{t}, \bar{X}, \bar{S})$ (recall that \bar{X} was assumed to be strictly primal feasible), the most natural, from the viewpoint of Computational Mathematics, way to achieve our target is as follows:

1. We choose somehow a desired new value $t_+ > \bar{t}$ of the penalty parameter;
2. We linearize the left hand side $G_{t_+}(X, S)$ of the system of nonlinear equations (4.5.2) at the point (\bar{X}, \bar{S}) , and replace (4.5.2) with the linearized system of equations

$$G_{t_+}(\bar{X}, \bar{S}) + \frac{\partial G_{t_+}(\bar{X}, \bar{S})}{\partial X}(X - \bar{X}) + \frac{\partial G_{t_+}(\bar{X}, \bar{S})}{\partial S}(S - \bar{S}) = 0 \quad (4.5.3)$$

3. We define the corrections ΔX , ΔS from the requirement that the updated pair $X_+ = \bar{X} + \Delta X$, $S_+ = \bar{S} + \Delta S$ must satisfy (4.5.1) and the linearized version (4.5.3) of (4.5.2). In other words, the corrections should solve the system

$$\begin{aligned} \Delta X &\in \mathcal{L}, \\ \Delta S &\in \mathcal{L}^\perp, \\ G_{t_+}(\bar{X}, \bar{S}) + \frac{\partial G_{t_+}(\bar{X}, \bar{S})}{\partial X} \Delta X + \frac{\partial G_{t_+}(\bar{X}, \bar{S})}{\partial S} \Delta S &= 0 \end{aligned} \quad (4.5.4)$$

4. Finally, we define X_+ and S_+ as

$$\begin{aligned} X_+ &= \bar{X} + \Delta X, \\ S_+ &= \bar{S} + \Delta S. \end{aligned} \quad (4.5.5)$$

¹²⁾ Of course, besides knowing how to trace the central path, we should also know how to initialize this process – how to come close to the path to be able to start its tracing. There are different techniques to resolve this “initialization difficulty”, and basically all of them achieve the goal by using the same path-tracing technique, now applied to an appropriate auxiliary problem where the “initialization difficulty” does not arise at all. Thus, at the level of ideas the initialization techniques do not add something essentially new, which allows us to skip in our presentation all initialization-related issues.

The primal-dual IP methods we are describing basically fit the outlined scheme, up to the following two important points:

- If the current iterate (\bar{X}, \bar{S}) is not enough close to $Z_*(\bar{t})$, and/or if the desired improvement $t_+ - \bar{t}$ is too large, the corrections given by the outlined scheme may be too large; as a result, the updating (4.5.5) as it is may be inappropriate, e.g., X_+ , or S_+ , or both, may be kicked out of the cone \mathbf{K} . (Why not: linearized system (4.5.3) approximates well the “true” system (4.5.2) only locally, and we have no reasons to trust in corrections coming from the linearized system, when these corrections are large.)

There is a standard way to overcome the outlined difficulty – to use the corrections in a damped fashion, namely, to replace the updating (4.5.5) with

$$\begin{aligned} X_+ &= \bar{X} + \alpha \Delta X, \\ S_+ &= \bar{S} + \beta \Delta S, \end{aligned} \tag{4.5.6}$$

and to choose the stepsizes $\alpha > 0, \beta > 0$ from additional “safety” considerations, like ensuring the updated pair (X_+, S_+) to reside in the interior of \mathbf{K} , or enforcing it to stay in a desired neighbourhood of the central path, or whatever else. In IP methods, the solution $(\Delta X, \Delta S)$ of (4.5.4) plays the role of *search direction* (and this is how it is called), and the actual corrections are proportional to the search ones rather than to be exactly the same. In this sense the situation is completely similar to the one with the Newton method from Section 4.2.2 (which is natural: the latter method is *exactly* the linearization method for solving the Fermat equation $\nabla f(x) = 0$).

- The “augmented complementary slackness” system (4.5.2) can be written down in many different forms which are equivalent to each other in the sense that they share a common solution set. E.g., we have the same reasons to express the augmented complementary slackness requirement by the nonlinear system (4.5.2) as to express it by the system

$$\hat{G}_t(X, S) \equiv X + t^{-1} \nabla K(S) = 0,$$

not speaking about other possibilities. And although all systems of nonlinear equations

$$H_t(X, S) = 0$$

expressing the augmented complementary slackness are “equivalent” in the sense that they share a common solution set, their linearizations are different and thus – lead to different search directions and finally to different path-following methods. Choosing appropriate (in general even varying from iteration to iteration) analytic representation of the augmented complementary slackness requirement, one can gain a lot in the performance of the resulting path-following method, and the IP machinery facilitates this flexibility (see “SDP case examples” below).

4.5.2 Speed of path-tracing

In the LP-CQP-SDP situation, the speed at which the best, from the theoretical viewpoint, path-following methods manage to trace the path, is inverse proportional to the square root of the parameter $\theta(K)$ of the underlying canonical barrier. It means the following. Started at a point (t^0, X^0, S^0) from the neighbourhood $\mathcal{N}_{0.1}$ of the central path, the method after $O(1)\sqrt{\theta(K)}$ steps

reaches the point ($t^1 = 2t^0, X^1, S^1$) from the same neighbourhood, after the same $O(1)\sqrt{\theta(K)}$ steps more reaches the point ($t^2 = 2^2t^0, X^2, S^2$) from the neighbourhood, and so on – it takes the method a *fixed number* $O(1)\sqrt{\theta(K)}$ steps to increase by factor 2 the current value of the penalty parameter, staying all the time in $\mathcal{N}_{0.1}$. By (4.4.7) it means that every $O(1)\sqrt{\theta(K)}$ steps of the method reduce the (upper bound on the) inaccuracy of current approximate solutions by factor 2, or, which is the same, add a fixed number of accuracy digits to these solutions. Thus, “the cost of an accuracy digit” for the (best) path-following methods is $O(1)\sqrt{\theta(K)}$ steps. To realize what this indeed mean, we should, of course, know how “heavy” a step is – what is its arithmetic cost. Well, the arithmetic cost of a step for the “cheapest among the fastest” IP methods as applied to (CP) is as if all operations carried out at a step were those required by

1. Assembling, given a point $X \in \text{int } \mathbf{K}$, the symmetric $n \times n$ matrix ($n = \dim x$)

$$\mathcal{H} = \mathcal{A}^*[\nabla^2 K(X)]\mathcal{A};$$

2. Subsequent Choleski factorization of the matrix \mathcal{H} (which, due to its origin, is symmetric positive definite and thus admits Choleski decomposition $\mathcal{H} = DD^T$ with lower triangular D).

Looking at (Cone), (CP) and (4.3.1), we immediately conclude that the arithmetic cost of assembling and factorizing \mathcal{H} is polynomial in the size $\dim \text{Data}(\cdot)$ of the data defining (CP), and that the parameter $\theta(K)$ also is polynomial in this size. Thus, the cost of an accuracy digit for the methods in question is polynomial in the size of the data, as is required from polynomial time methods¹³⁾. Explicit complexity bounds for \mathcal{LP}_b , \mathcal{CQP}_b , \mathcal{SDP}_b are given in Sections 4.6.1, 4.6.2, 4.6.3, respectively.

4.5.3 The primal and the dual path-following methods

The simplest way to implement the path-following scheme from Section 4.5.1 is to linearize the augmented complementary slackness equations (4.5.2) as they are, ignoring the option to rewrite these equations equivalently before linearization. Let us look at the resulting method in more details. Linearizing (4.5.2) at a current iterate \bar{X}, \bar{S} , we get the vector equation

$$t_+(\bar{S} + \Delta S) + \nabla K(\bar{X}) + [\nabla^2 K(\bar{X})]\Delta X = 0,$$

where t_+ is the target value of the penalty parameter. The system (4.5.4) now becomes

$$\begin{array}{ll}
 (a) & \Delta X \in \mathcal{L} \\
 & \Downarrow \\
 (a') & \Delta X = \mathcal{A}\Delta x \quad [\Delta x \in \mathbf{R}^n] \\
 (b) & \Delta S \in \mathcal{L}^\perp \\
 & \Downarrow \\
 (b') & \mathcal{A}^*\Delta S = 0 \\
 (c) & t_+[\bar{S} + \Delta S] + \nabla K(\bar{X}) + [\nabla^2 K(\bar{X})]\Delta X = 0;
 \end{array} \tag{4.5.7}$$

¹³⁾ Strictly speaking, the outlined complexity considerations are applicable to the “highway” phase of the solution process, after we once have reached the neighbourhood $\mathcal{N}_{0.1}$ of the central path. However, the results of our considerations remain unchanged after the initialization expenses are taken into account, see Section 4.6.

the unknowns here are ΔX , ΔS and Δx . To process the system, we eliminate ΔX via (a') and multiply both sides of (c) by \mathcal{A}^* , thus getting the equation

$$\underbrace{\mathcal{A}^*[\nabla^2 K(\bar{X})]\mathcal{A}}_{\mathcal{H}} \Delta x + [t_+ \mathcal{A}^*[\bar{S} + \Delta S] + \mathcal{A}^* \nabla K(\bar{X})] = 0. \quad (4.5.8)$$

Note that $\mathcal{A}^*[\bar{S} + \Delta S] = c$ is the objective of (CP) (indeed, $\bar{S} \in \mathcal{L}^\perp + C$, i.e., $\mathcal{A}^*\bar{S} = c$, while $\mathcal{A}^*\Delta S = 0$ by (b')). Consequently, (4.5.8) becomes the *primal Newton system*

$$\mathcal{H}\Delta x = -[t_+ c + \mathcal{A}^* \nabla K(\bar{X})]. \quad (4.5.9)$$

Solving this system (which is possible – it is easily seen that the $n \times n$ matrix \mathcal{H} is positive definite), we get Δx and then set

$$\begin{aligned} \Delta X &= \mathcal{A}\Delta x, \\ \Delta S &= -t_+^{-1}[\nabla K(\bar{X}) + [\nabla^2 K(\bar{X})\Delta X] - \bar{S}], \end{aligned} \quad (4.5.10)$$

thus getting a solution to (4.5.7). Restricting ourselves with the stepsizes $\alpha = \beta = 1$ (see (4.5.6)), we come to the “closed form” description of the method:

$$\begin{aligned} (a) \quad & t \mapsto t_+ > t \\ (b) \quad x \mapsto x_+ &= x + \underbrace{(-[\mathcal{A}^*(\nabla^2 K(X))\mathcal{A}]^{-1}[t_+ c + \mathcal{A}^* \nabla K(X)])}_{\Delta x}, \\ (c) \quad S \mapsto S_+ &= -t_+^{-1}[\nabla K(X) + [\nabla^2 K(X)]\mathcal{A}\Delta x], \end{aligned} \quad (4.5.11)$$

where x is the current iterate in the space \mathbf{R}^n of design variables and $X = \mathcal{A}x - B$ is its image in the space E .

The resulting scheme admits a quite natural explanation. Consider the function

$$F(x) = K(\mathcal{A}x - B);$$

you can immediately verify that this function is a barrier for the feasible set of (CP). Let also

$$F_t(x) = tc^T x + F(x)$$

be the associated barrier-generated family of penalized objectives. Relation (4.5.11.b) says that the iterates in the space of design variables are updated according to

$$x \mapsto x_+ = x - [\nabla^2 F_{t_+}(x)]^{-1} \nabla F_{t_+}(x),$$

i.e., the process in the space of design variables is exactly the process (4.2.1) from Section 4.2.3.

Note that (4.5.11) is, essentially, a purely primal process (this is where the name of the method comes from). Indeed, the dual iterates S, S_+ just do not appear in formulas for x_+, X_+ , and in fact the dual solutions are no more than “shadows” of the primal ones.

Remark 4.5.1 When constructing the primal path-following method, we have started with the augmented slackness equations in form (4.5.2). Needless to say, we could start our developments with the same conditions written down in the “swapped” form

$$X + t^{-1} \nabla K(S) = 0$$

as well, thus coming to what is called “dual path-following method”. Of course, as applied to a given pair (P), (D), the dual path-following method differs from the primal one. However, the constructions and results related to the dual path-following method require no special care – they can be obtained from their “primal counterparts” just by swapping “primal” and “dual” entities.

The complexity analysis of the primal path-following method can be summarized in the following

Theorem 4.5.1 *Let $0 < \chi \leq \kappa \leq 0.1$. Assume that we are given a starting point (t_0, x_0, S_0) such that $t_0 > 0$ and the point*

$$(X_0 = \mathcal{A}x_0 - B, S_0)$$

is κ -close to $Z_(t_0)$:*

$$\text{dist}((X_0, S_0), Z_*(t_0)) \leq \kappa.$$

Starting with (t_0, x_0, X_0, S_0) , let us iterate process (4.5.11) equipped with the penalty updating policy

$$t_+ = \left(1 + \frac{\chi}{\sqrt{\theta(K)}}\right) t \quad (4.5.12)$$

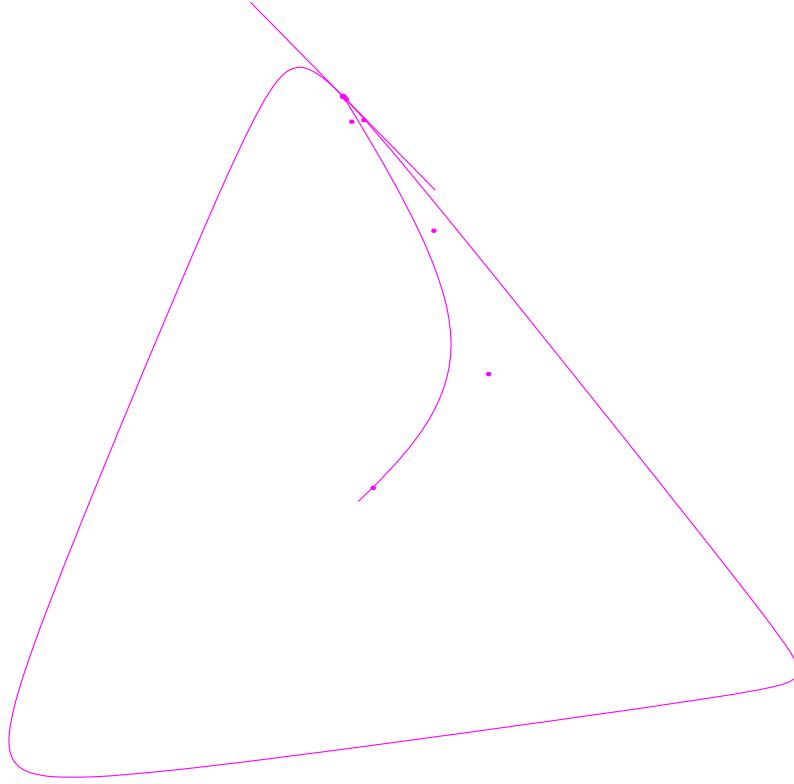
i.e., let us build the iterates (t_i, x_i, X_i, S_i) according to

$$\begin{aligned} t_i &= \left(1 + \frac{\chi}{\sqrt{\theta(K)}}\right) t_{i-1}, \\ x_i &= x_{i-1} - \underbrace{[\mathcal{A}^*(\nabla^2 K(X_{i-1}))\mathcal{A}]^{-1}[t_i c + \mathcal{A}^*\nabla K(X_{i-1})]}_{\Delta x_i}, \\ X_i &= \mathcal{A}x_i - B, \\ S_i &= -t_i^{-1}[\nabla K(X_{i-1}) + [\nabla^2 K(X_{i-1})]\mathcal{A}\Delta x_i] \end{aligned}$$

The resulting process is well-defined and generates strictly primal-dual feasible pairs (X_i, S_i) such that (t_i, X_i, S_i) stay in the neighbourhood \mathcal{N}_κ of the primal-dual central path.

The theorem says that with properly chosen κ, χ (e.g., $\kappa = \chi = 0.1$) we can, getting once close to the primal-dual central path, trace it by the primal path-following method, keeping the iterates in \mathcal{N}_κ -neighbourhood of the path and increasing the penalty parameter by an absolute constant factor every $O(\sqrt{\theta(K)})$ steps – exactly as it was claimed in Sections 4.2.3, 4.5.2. This fact is extremely important theoretically; in particular, it underlies the polynomial time complexity bounds for LP, CQP and SDP from Section 4.6 below. As a practical tool, the primal and the dual path-following methods, at least in their short-step form presented above, are not that attractive. The computational power of the methods can be improved by passing to appropriate large-step versions of the algorithms, but even these versions are thought of to be inferior as compared to “true” primal-dual path-following methods (those which “indeed work with both (P) and (D)”, see below). There are, however, cases when the primal or the dual path-following scheme seems to be unavoidable; these are, essentially, the situations where the pair (P), (D) is “highly asymmetric”, e.g., (P) and (D) have different by order of magnitudes design dimensions $\dim \mathcal{L}$, $\dim \mathcal{L}^\perp$. Here it becomes too expensive computationally to treat (P), (D) in a “nearly symmetric way”, and it is better to focus solely on the problem with smaller design dimension.

To get an impression of how the primal path-following method works, here is a picture:



What you see is the 2D feasible set of a toy SDP ($\mathbf{K} = \mathbf{S}_+^3$). “Continuous curve” is the primal central path; dots are iterates x_i of the algorithm. We cannot draw the dual solutions, since they “live” in 4-dimensional space ($\dim \mathcal{L}^\perp = \dim \mathbf{S}^3 - \dim \mathcal{L} = 6 - 2 = 4$)

Here are the corresponding numbers:

Itr#	Objective	Duality Gap	Itr#	Objective	Duality Gap
1	-0.100000	2.96	7	-1.359870	8.4e-4
2	-0.906963	0.51	8	-1.360259	2.1e-4
3	-1.212689	0.19	9	-1.360374	5.3e-5
4	-1.301082	6.9e-2	10	-1.360397	1.4e-5
5	-1.349584	2.1e-2	11	-1.360404	3.8e-6
6	-1.356463	4.7e-3	12	-1.360406	9.5e-7

4.5.4 The SDP case

In what follows, we specialize the primal-dual path-following scheme in the SDP case and carry out its complexity analysis.

4.5.4.1 The path-following scheme in SDP

Let us look at the outlined scheme in the SDP case. Here the system of nonlinear equations (4.5.2) becomes (see (4.3.1))

$$G_t(X, S) \equiv S - t^{-1}X^{-1} = 0, \quad (4.5.13)$$

X, S being positive definite $k \times k$ symmetric matrices.

Recall that our generic scheme of a path-following IP method suggests, given a current triple $(\bar{t}, \bar{X}, \bar{S})$ with positive \bar{t} and strictly primal, respectively, dual feasible \bar{X} and \bar{S} , to update this triple into a new triple (t_+, X_+, S_+) of the same type as follows:

(i) First, we somehow rewrite the system (4.5.13) as an equivalent system

$$\bar{G}_t(X, S) = 0; \quad (4.5.14)$$

(ii) Second, we choose somehow a new value $t_+ > \bar{t}$ of the penalty parameter and linearize system (4.5.14) (with t set to t_+) at the point (\bar{X}, \bar{S}) , thus coming to the system of linear equations

$$\frac{\partial \bar{G}_{t_+}(\bar{X}, \bar{S})}{\partial X} \Delta X + \frac{\partial \bar{G}_{t_+}(\bar{X}, \bar{S})}{\partial S} \Delta S = -\bar{G}_{t_+}(\bar{X}, \bar{S}), \quad (4.5.15)$$

for the “corrections” $(\Delta X, \Delta S)$;

We add to (4.5.15) the system of linear equations on $\Delta X, \Delta S$ expressing the requirement that a shift of (\bar{X}, \bar{S}) in the direction $(\Delta X, \Delta S)$ should preserve the validity of the linear constraints in (P), (D), i.e., the equations saying that $\Delta X \in \mathcal{L}$, $\Delta S \in \mathcal{L}^\perp$. These linear equations can be written down as

$$\begin{aligned} \Delta X &= \mathcal{A}\Delta x & [\Leftrightarrow \Delta X \in \mathcal{L}] \\ \mathcal{A}^*\Delta S &= 0 & [\Leftrightarrow \Delta S \in \mathcal{L}^\perp] \end{aligned} \quad (4.5.16)$$

(iii) We solve the system of linear equations (4.5.15), (4.5.16), thus obtaining a primal-dual search direction $(\Delta X, \Delta S)$, and update current iterates according to

$$X_+ = \bar{X} + \alpha \Delta x, \quad S_+ = \bar{S} + \beta \Delta S$$

where the primal and the dual stepsizes α, β are given by certain “side requirements”.

The major “degree of freedom” of the construction comes from (i) – from how we construct the system (4.5.14). A very popular way to handle (i), the way which indeed leads to *primal-dual* methods, starts from rewriting (4.5.13) in a form symmetric w.r.t. X and S . To this end we first observe that (4.5.13) is equivalent to every one of the following two matrix equations:

$$XS = t^{-1}I; \quad SX = t^{-1}I.$$

Adding these equations, we get a “symmetric” w.r.t. X, S matrix equation

$$XS + SX = 2t^{-1}I, \quad (4.5.17)$$

which, by its origin, is a consequence of (4.5.13). On a closest inspection, it turns out that (4.5.17), regarded as a matrix equation with *positive definite* symmetric matrices, is equivalent to (4.5.13). It is possible to use in the role of (4.5.14) the matrix equation (4.5.17) as it is; this policy leads to the so called AHO (Alizadeh-Overton-Haeberly) search direction and the “ $XS + SX$ ” primal-dual path-following method.

It is also possible to use a “scaled” version of (4.5.17). Namely, let us choose somehow a positive definite *scaling matrix* Q and observe that our original matrix equation (4.5.13) says that $S = t^{-1}X^{-1}$, which is exactly the same as to say that $Q^{-1}SQ^{-1} = t^{-1}(QXQ)^{-1}$; the latter, in turn, is equivalent to every one of the matrix equations

$$QXSQ^{-1} = t^{-1}I; \quad Q^{-1}SXQ = t^{-1}I;$$

Adding these equations, we get the scaled version of (4.5.17):

$$QXSQ^{-1} + Q^{-1}SXQ = 2t^{-1}I, \quad (4.5.18)$$

which, same as (4.5.17) itself, is equivalent to (4.5.13).

With (4.5.18) playing the role of (4.5.14), we get a quite flexible scheme with a huge freedom for choosing the scaling matrix Q , which in particular can be varied from iteration to iteration. As we shall see in a while, this freedom reflects the intrinsic (and extremely important in the interior-point context) symmetries of the semidefinite cone.

Analysis of the path-following methods based on search directions coming from (4.5.18) (“Zhang’s family of search directions”) simplifies a lot when at every iteration we choose its own scaling matrix and ensure that the matrices

$$\tilde{S} = Q^{-1}\bar{S}Q^{-1}, \quad \hat{X} = Q\bar{X}Q$$

commute (\bar{X}, \bar{S} are the iterates to be updated); we call such a policy a “commutative scaling”. Popular commutative scalings are:

1. $Q = \bar{S}^{1/2}$ ($\tilde{S} = I, \hat{X} = \bar{S}^{1/2}\bar{X}\bar{S}^{1/2}$) (the “ XS ” method);
2. $Q = \bar{X}^{-1/2}$ ($\tilde{S} = \bar{X}^{1/2}\bar{S}\bar{X}^{1/2}, \hat{X} = I$) (the “ SX ” method);
3. Q is such that $\tilde{S} = \hat{X}$ (the NT (Nesterov-Todd) method, extremely attractive and deep)

If \bar{X} and \bar{S} were just positive reals, the formula for Q would be simple: $Q = \left(\frac{\bar{S}}{\bar{X}}\right)^{1/4}$.

In the matrix case this simple formula becomes a bit more complicated (to make our life easier, below we write X instead of \bar{X} and S instead of \bar{S}):

$$Q = P^{1/2}, \quad P = X^{-1/2}(X^{1/2}SX^{1/2})^{-1/2}X^{1/2}S.$$

We should verify that (a) P is symmetric positive definite, so that Q is well-defined, and that (b) $Q^{-1}SQ^{-1} = QXQ$.

(a): Let us first verify that P is symmetric:

$$\begin{aligned} P &=? P^T \\ &\Updownarrow \\ X^{-1/2}(X^{1/2}SX^{1/2})^{-1/2}X^{1/2}S &=? SX^{1/2}(X^{1/2}SX^{1/2})^{-1/2}X^{-1/2} \\ &\Updownarrow \\ (X^{-1/2}(X^{1/2}SX^{1/2})^{-1/2}X^{1/2}S) &(X^{1/2}(X^{1/2}SX^{1/2})^{1/2}X^{-1/2}S^{-1}) ?=? I \\ &\Updownarrow \\ X^{-1/2}(X^{1/2}SX^{1/2})^{-1/2}(X^{1/2}SX^{1/2})(X^{1/2}SX^{1/2})^{1/2}X^{-1/2}S^{-1} &=? I \\ &\Updownarrow \\ X^{-1/2}(X^{1/2}SX^{1/2})X^{-1/2}S^{-1} &=? I \end{aligned}$$

and the concluding $?=?$ indeed is $=$.

Now let us verify that P is positive definite. Recall that the spectrum of the product of two square matrices, symmetric or not, remains unchanged when swapping the factors. Therefore, denoting $\sigma(A)$ the spectrum of A , we have

$$\begin{aligned} \sigma(P) &= \sigma(X^{-1/2}(X^{1/2}SX^{1/2})^{-1/2}X^{1/2}S) \\ &= \sigma((X^{1/2}SX^{1/2})^{-1/2}X^{1/2}SX^{-1/2}) \\ &= \sigma((X^{1/2}SX^{1/2})^{-1/2}(X^{1/2}SX^{1/2})X^{-1}) \\ &= \sigma((X^{1/2}SX^{1/2})^{1/2}X^{-1}) \\ &= \sigma(X^{-1/2}(X^{1/2}SX^{1/2})^{1/2}X^{-1/2}), \end{aligned}$$

and the argument of the concluding $\sigma(\cdot)$ clearly is a positive definite symmetric matrix. Thus, the spectrum of *symmetric* matrix P is positive, i.e., P is positive definite.

(b): To verify that $QXQ = Q^{-1}SQ^{-1}$, i.e., that $P^{1/2}XP^{1/2} = P^{-1/2}SP^{-1/2}$, is the same as to verify that $PXP = S$. The latter equality is given by the following computation:

$$\begin{aligned} PXP &= (X^{-1/2}(X^{1/2}SX^{1/2})^{-1/2}X^{1/2}S)X(X^{-1/2}(X^{1/2}SX^{1/2})^{-1/2}X^{1/2}S) \\ &= X^{-1/2}(X^{1/2}SX^{1/2})^{-1/2}(X^{1/2}SX^{1/2})(X^{1/2}SX^{1/2})^{-1/2}X^{1/2}S \\ &= X^{-1/2}X^{1/2}S \\ &= S. \end{aligned}$$

You should not think that Nesterov and Todd guessed the formula for this scaling matrix. They did much more: they have developed an extremely deep theory (covering the general LP-CQP-SDP case, not just the SDP one!) which, among other things, guarantees that the desired scaling matrix exists (and even is unique). After the existence is established, it becomes much easier (although still not that easy) to find an explicit formula for Q .

4.5.4.2 Complexity analysis

We are about to carry out the complexity analysis of the primal-dual path-following methods based on “commutative” Zhang’s scalings. This analysis, although not that difficult, is more technical than whatever else in our course, and a non-interested reader may skip it without any harm.

Scalings. We already have mentioned what a *scaling* of \mathbf{S}_+^k is: this is the linear one-to-one transformation of \mathbf{S}^k given by the formula

$$H \mapsto QHQ^T, \quad (\text{Scl})$$

where Q is a nonsingular *scaling matrix*. It is immediately seen that (Scl) is a symmetry of the semidefinite cone \mathbf{S}_+^k – it maps the cone onto itself. This family of symmetries is quite rich: for every pair of points A, B from the interior of the cone, there exists a scaling which maps A onto B , e.g., the scaling

$$H \mapsto \underbrace{(B^{1/2}A^{-1/2})}_Q H \underbrace{(A^{-1/2}B^{1/2})}_{Q^T}.$$

Essentially, this is exactly the existence of that rich family of symmetries of the underlying cones which makes SDP (same as LP and CQP, where the cones also are “perfectly symmetric”) especially well suited for IP methods.

In what follows we will be interested in scalings associated with *positive definite* scaling matrices. The scaling given by such a matrix Q (X, S, \dots) will be denoted by \mathcal{Q} (resp., $\mathcal{X}, \mathcal{S}, \dots$):

$$\mathcal{Q}[H] = QHQ.$$

Given a problem of interest (CP) (where $\mathbf{K} = \mathbf{S}_+^k$) and a scaling matrix $Q \succ 0$, we can *scale* the problem, i.e., pass from it to the problem

$$\min_x \{c^T x : \mathcal{Q}[\mathcal{A}x - \mathcal{B}] \succeq 0\} \quad \mathcal{Q}(\text{CP})$$

which, of course, is equivalent to (CP) (since $\mathcal{Q}[H]$ is positive semidefinite iff H is so). In terms of “geometric reformulation” (P) of (CP), this transformation is nothing but the substitution of variables

$$QXQ = Y \Leftrightarrow X = Q^{-1}YQ^{-1};$$

with respect to Y -variables, (P) is the problem

$$\min_Y \{\text{Tr}(C[Q^{-1}YQ^{-1}]) : Y \in \mathcal{Q}(\mathcal{L}) - \mathcal{Q}[\mathcal{B}], Y \succeq 0\},$$

i.e., the problem

$$\begin{aligned} & \min_Y \left\{ \text{Tr}(\tilde{C}Y) : Y \in \hat{\mathcal{L}} - \hat{B}, Y \succeq 0 \right\} \\ & \left[\tilde{C} = Q^{-1}CQ^{-1}, \hat{B} = QBQ, \hat{\mathcal{L}} = \text{Im}(\mathcal{QA}) = \mathcal{Q}(\mathcal{L}) \right] \end{aligned} \quad (\hat{P})$$

The problem dual to (\hat{P}) is

$$\max_Z \left\{ \text{Tr}(\hat{B}Z) : Z \in \hat{\mathcal{L}}^\perp + \hat{C}, Z \succeq 0 \right\}. \quad (\tilde{D})$$

It is immediate to realize what is $\hat{\mathcal{L}}^\perp$:

$$\langle Z, QXQ \rangle_F = \text{Tr}(ZQXQ) = \text{Tr}(QZQX) = \langle QZQ, X \rangle_F;$$

thus, Z is orthogonal to every matrix from $\hat{\mathcal{L}}$, i.e., to every matrix of the form QXQ with $X \in \mathcal{L}$ iff the matrix QZQ is orthogonal to every matrix from \mathcal{L} , i.e., iff $QZQ \in \mathcal{L}^\perp$. It follows that

$$\hat{\mathcal{L}}^\perp = \mathcal{Q}^{-1}(\mathcal{L}^\perp).$$

Thus, when acting on the primal-dual pair $(P), (D)$ of SDP's, a scaling, given by a matrix $Q \succ 0$, converts it into another primal-dual pair of problems, and this new pair is as follows:

- The “primal” geometric data – the subspace \mathcal{L} and the primal shift B (which has a part-time job to be the dual objective as well) – are replaced with their images under the mapping \mathcal{Q} ;
- The “dual” geometric data – the subspace \mathcal{L}^\perp and the dual shift C (it is the primal objective as well) – are replaced with their images under the mapping \mathcal{Q}^{-1} inverse to \mathcal{Q} ; this inverse mapping again is a scaling, the scaling matrix being Q^{-1} .

We see that it makes sense to speak about *primal-dual scaling* which acts on both the primal and the dual variables and maps a primal variable X onto QXQ , and a dual variable S onto $Q^{-1}SQ^{-1}$. Formally speaking, the primal-dual scaling associated with a matrix $Q \succ 0$ is the linear transformation $(X, S) \mapsto (QXQ, Q^{-1}SQ^{-1})$ of the direct product of two copies of \mathbf{S}^k (the “primal” and the “dual” ones). A primal-dual scaling acts naturally on different entities associated with a primal-dual pair $(P), (S)$, in particular, at:

- the pair $(P), (D)$ itself – it is converted into another primal-dual pair of problems $(\hat{P}), (\tilde{D})$;
- a primal-dual feasible pair (X, S) of solutions to $(P), (D)$ – it is converted to the pair $(\hat{X} = QXQ, \tilde{S} = Q^{-1}SQ^{-1})$, which, as it is immediately seen, is a pair of feasible solutions to $(\hat{P}), (\tilde{D})$.

Note that the primal-dual scaling preserves strict feasibility and the duality gap:

$$\text{DualityGap}_{P,D}(X, S) = \text{Tr}(XS) = \text{Tr}(QXSQ^{-1}) = \text{Tr}(\hat{X}\tilde{S}) = \text{DualityGap}_{\hat{P},\tilde{D}}(\hat{X}, \tilde{S});$$

- the primal-dual central path $(X_*(\cdot), S_*(\cdot))$ of $(P), (D)$; it is converted into the curve $(\hat{X}_*(t) = QX_*(t)Q, \tilde{S}_*(t) = Q^{-1}S_*(t)Q^{-1})$, which is nothing but the primal-dual central path $\bar{Z}(t)$ of the primal-dual pair $(\hat{P}), (\tilde{D})$.

The latter fact can be easily derived from the characterization of the primal-dual central path; a more instructive derivation is based on the fact that our “hero” – the barrier $S_k(\cdot)$ – is “semi-invariant” w.r.t. scaling:

$$S_k(Q(X)) = -\ln \text{Det}(QXQ) = -\ln \text{Det}(X) - 2\ln \text{Det}(Q) = S_k(X) + \text{const}(Q).$$

Now, a point on the primal central path of the problem (\hat{P}) associated with penalty parameter t , let this point be temporarily denoted by $Y(t)$, is the unique minimizer of the aggregate

$$S_k^t(Y) = t\langle Q^{-1}CQ^{-1}, Y \rangle_F + S_k(Y) \equiv t\text{Tr}(Q^{-1}CQ^{-1}Y) + S_k(Y)$$

over the set of strictly feasible solutions of (\hat{P}) . The latter set is exactly the image of the set of strictly feasible solutions of (P) under the transformation \mathcal{Q} , so that $Y(t)$ is the image, under the same transformation, of the point, let it be called $X(t)$, which minimizes the aggregate

$$S_k^t(QXQ) = t\text{Tr}((Q^{-1}CQ^{-1})(QXQ)) + S_k(QXQ) = t\text{Tr}(CX) + S_k(X) + \text{const}(Q)$$

over the set of strictly feasible solutions to (P). We see that $X(t)$ is exactly the point $X_*(t)$ on the primal central path associated with problem (P). Thus, the point $Y(t)$ of the primal central path associated with (\widehat{P}) is nothing but $\widehat{X}_*(t) = QX_*(t)Q$. Similarly, the point of the central path associated with the problem (\widetilde{D}) is exactly $\widetilde{S}_*(t) = Q^{-1}S_*(t)Q^{-1}$.

- the neighbourhood \mathcal{N}_κ of the primal-dual central path $Z(\cdot)$ associated with the pair of problems (P), (D) (see (4.4.8)). As you can guess, the image of \mathcal{N}_κ is exactly the neighbourhood $\overline{\mathcal{N}}_\kappa$, given by (4.4.8), of the primal-dual central path $\overline{Z}(\cdot)$ of (\widehat{P}) , (\widetilde{D}) .

The latter fact is immediate: for a pair (X, S) of strictly feasible primal and dual solutions to (P), (D) and a $t > 0$ we have (see (4.4.6)):

$$\begin{aligned} \text{dist}^2((\widehat{X}, \widetilde{S}), \overline{Z}_*(t)) &= \text{Tr}([QXQ](tQ^{-1}SQ^{-1} - [QXQ]^{-1})[QXQ](tQ^{-1}SQ^{-1} - [QXQ]^{-1})) \\ &= \text{Tr}(QX(tS - X^{-1})X(tS - X^{-1})Q^{-1}) \\ &= \text{Tr}(X(tS - X^{-1})X(tS - X^{-1})) \\ &= \text{dist}^2((X, S), Z_*(t)). \end{aligned}$$

Primal-dual short-step path-following methods based on commutative scalings. Path-following methods we are about to consider trace the primal-dual central path of (P), (D), staying in \mathcal{N}_κ -neighbourhood of the path; here $\kappa \leq 0.1$ is fixed. The path is traced by iterating the following updating:

(U): Given a current pair of strictly feasible primal and dual solutions (\bar{X}, \bar{S}) such that the triple

$$\left(\bar{t} = \frac{k}{\text{Tr}(\bar{X}\bar{S})}, \bar{X}, \bar{S} \right) \quad (4.5.19)$$

belongs to \mathcal{N}_κ , i.e. (see (4.4.6))

$$\|\bar{t}\bar{X}^{1/2}\bar{S}\bar{X}^{1/2} - I\|_2 \leq \kappa, \quad (4.5.20)$$

we

1. Choose the new value t_+ of the penalty parameter according to

$$t_+ = \left(1 - \frac{\chi}{\sqrt{k}} \right)^{-1} \bar{t}, \quad (4.5.21)$$

where $\chi \in (0, 1)$ is a parameter of the method;

2. Choose somehow the scaling matrix $Q \succ 0$ such that the matrices $\widehat{X} = Q\bar{X}Q$ and $\widetilde{S} = Q^{-1}\bar{S}Q^{-1}$ commute with each other;
3. Linearize the equation

$$QXSQ^{-1} + Q^{-1}SXQ = \frac{2}{t_+}I$$

at the point (\bar{X}, \bar{S}) , thus coming to the equation

$$Q[\Delta XS + X\Delta S]Q^{-1} + Q^{-1}[\Delta SX + S\Delta X]Q = \frac{2}{t_+}I - [Q\bar{X}\bar{S}Q^{-1} + Q^{-1}\bar{S}\bar{X}Q]; \quad (4.5.22)$$

4. Add to (4.5.22) the linear equations

$$\begin{aligned} \Delta X &\in \mathcal{L}, \\ \Delta S &\in \mathcal{L}^\perp; \end{aligned} \quad (4.5.23)$$

5. Solve system (4.5.22), (4.5.23), thus getting “primal-dual search direction” $(\Delta X, \Delta S)$;

6. Update current primal-dual solutions (\bar{X}, \bar{S}) into a new pair (X_+, S_+) according to

$$X_+ = \bar{X} + \Delta X, \quad S_+ = \bar{S} + \Delta S.$$

We already have explained the ideas underlying (U), up to the fact that in our previous explanations we dealt with three “independent” entities \bar{t} (current value of the penalty parameter), \bar{X}, \bar{S} (current primal and dual solutions), while in (U) \bar{t} is a function of \bar{X}, \bar{S} :

$$\bar{t} = \frac{k}{\text{Tr}(\bar{X}\bar{S})}. \quad (4.5.24)$$

The reason for establishing this dependence is very simple: if (t, X, S) were on the primal-dual central path: $XS = t^{-1}I$, then, taking traces, we indeed would get $t = \frac{k}{\text{Tr}(XS)}$. Thus, (4.5.24) is a reasonable way to reduce the number of “independent entities” we deal with.

Note also that (U) is a “pure Newton scheme” – here the primal and the dual stepsizes are equal to 1 (cf. (4.5.6)).

The major element of the complexity analysis of path-following polynomial time methods for SDP is as follows:

Theorem 4.5.2 *Let the parameters κ, χ of (U) satisfy the relations*

$$0 < \chi \leq \kappa \leq 0.1. \quad (4.5.25)$$

Let, further, (\bar{X}, \bar{S}) be a pair of strictly feasible primal and dual solutions to (P), (D) such that the triple (4.5.19) satisfies (4.5.20). Then the updated pair (X_+, S_+) is well-defined (i.e., system (4.5.22), (4.5.23) is solvable with a unique solution), X_+, S_+ are strictly feasible solutions to (P), (D), respectively,

$$t_+ = \frac{k}{\text{Tr}(X_+S_+)}$$

and the triple (t_+, X_+, S_+) belongs to \mathcal{N}_κ .

The theorem says that with properly chosen κ, χ (say, $\kappa = \chi = 0.1$), updating (U) converts a close to the primal-dual central path, in the sense of (4.5.20), strictly primal-dual feasible iterate (\bar{X}, \bar{S}) into a new strictly primal-dual feasible iterate with the same closeness-to-the-path property and larger, by factor $(1 + O(1)k^{-1/2})$, value of the penalty parameter. Thus, after we get close to the path – reach its 0.1-neighbourhood $\mathcal{N}_{0.1}$ – we are able to trace this path, staying in $\mathcal{N}_{0.1}$ and increasing the penalty parameter by absolute constant factor in $O(\sqrt{k}) = O(\sqrt{\theta(K)})$ steps, exactly as announced in Section 4.5.2.

Proof of Theorem 4.5.2. ¹⁰. Observe, first (this observation is crucial!) that it suffices to prove our Theorem in the particular case when \bar{X}, \bar{S} commute with each other and $Q = I$. Indeed, it is immediately seen that the updating (U) can be represented as follows:

1. We first scale by Q the “input data” of (U) – the primal-dual pair of problems (P), (D) and the strictly feasible pair \bar{X}, \bar{S} of primal and dual solutions to these problems, as explained in sect. “Scaling”. Note that the resulting entities – a pair of primal-dual problems and a strictly feasible pair of primal-dual solutions to these problems – are linked with each other exactly in the same fashion as the original entities, due to scaling invariance of the duality gap and the neighbourhood \mathcal{N}_κ . In addition, the scaled primal and dual solutions commute;
2. We apply to the “scaled input data” yielded by the previous step the updating (\hat{U}) completely similar to (U), but using the unit matrix in the role of Q ;
3. We “scale back” the result of the previous step, i.e., subject this result to the scaling associated with Q^{-1} , thus obtaining the updated iterate (X^+, S^+) .

Given that the second step of this procedure preserves primal-dual strict feasibility, w.r.t. the scaled primal-dual pair of problems, of the iterate and keeps the iterate in the κ -neighbourhood \mathcal{N}_κ of the corresponding central path, we could use once again the “scaling invariance” reasoning to assert that the result (X^+, S^+) of (U) is well-defined, is strictly feasible for (P), (D) and is close to the original central path, as claimed in the Theorem. Thus, all we need is to justify the above “Given”, and this is exactly the same as to prove the theorem in the particular case of $Q = I$ and commuting \bar{X}, \bar{S} . In the rest of the proof we assume that $Q = I$ and that the matrices \bar{X}, \bar{S} commute with each other. Due to the latter property, \bar{X}, \bar{S} are diagonal in a properly chosen orthonormal basis; representing all matrices from \mathbf{S}^k in this basis, we can reduce the situation to the case when \bar{X} and \bar{S} are diagonal. Thus, we may (and do) assume in the sequel that \bar{X} and \bar{S} are diagonal, with diagonal entries $x_i, s_i, i = 1, \dots, k$, respectively, and that $Q = I$. Finally, to simplify notation, we write t, X, S instead of $\bar{t}, \bar{X}, \bar{S}$, respectively.

2⁰. Our situation and goals now are as follows. We are given orthogonal to each other affine planes $\mathcal{L} - B, \mathcal{L}^\perp + C$ in \mathbf{S}^k and two positive definite diagonal matrices $X = \text{Diag}(\{x_i\}) \in \mathcal{L} - B, S = \text{Diag}(\{s_i\}) \in \mathcal{L}^\perp + C$. We set

$$\mu = \frac{1}{t} = \frac{\text{Tr}(XS)}{k}$$

and know that

$$\|tX^{1/2}SX^{1/2} - I\|_2 \leq \kappa.$$

We further set

$$\mu_+ = \frac{1}{t_+} = (1 - \chi k^{-1/2})\mu \quad (4.5.26)$$

and consider the system of equations w.r.t. unknown symmetric matrices $\Delta X, \Delta S$:

$$\begin{aligned} (a) \quad & \Delta X \in \mathcal{L} \\ (b) \quad & \Delta S \in \mathcal{L}^\perp \\ (c) \quad & \Delta XS + X\Delta S + \Delta SX + S\Delta X = 2\mu_+ I - 2XS \end{aligned} \quad (4.5.27)$$

We should prove that the system has a unique solution such that the matrices

$$X_+ = X + \Delta X, \quad S_+ = S + \Delta S$$

are

- (i) positive definite,
- (ii) belong, respectively, to $\mathcal{L} - B, \mathcal{L}^\perp + C$ and satisfy the relation

$$\text{Tr}(X_+ S_+) = \mu_+ k; \quad (4.5.28)$$

- (iii) satisfy the relation

$$\Omega \equiv \|\mu_+^{-1} X_+^{1/2} S_+ X_+^{1/2} - I\|_2 \leq \kappa. \quad (4.5.29)$$

Observe that the situation can be reduced to the one with $\mu = 1$. Indeed, let us pass from the matrices $X, S, \Delta X, \Delta S, X_+, S_+$ to $X, S' = \mu^{-1}S, \Delta X, \Delta S' = \mu^{-1}\Delta S, X_+, S'_+ = \mu^{-1}S_+$. Now the “we are given” part of our situation becomes as follows: we are given two diagonal positive definite matrices X, S' such that $X \in \mathcal{L} - B, S' \in \mathcal{L}^\perp + C', C' = \mu^{-1}C$,

$$\text{Tr}(XS') = k \times 1$$

and

$$\|X^{1/2}S'X^{1/2} - I\|_2 = \|\mu^{-1}X^{1/2}SX^{1/2} - I\|_2 \leq \kappa.$$

The “we should prove” part becomes: to verify that the system of equations

$$\begin{aligned} (a) \quad & \Delta X \in \mathcal{L} \\ (b) \quad & \Delta S' \in \mathcal{L}^\perp \\ (c) \quad & \Delta XS' + X\Delta S' + \Delta S'X + S'\Delta X = 2(1 - \chi k^{-1/2})I - 2XS' \end{aligned}$$

has a unique solution and that the matrices $X_+ = X + \Delta X$, $S'_+ = S' + \Delta S'_+$ are positive definite, are contained in $\mathcal{L} - B$, respectively, $\mathcal{L}^\perp + C'$ and satisfy the relations

$$\text{Tr}(X_+ S'_+) = \frac{\mu_+}{\mu} = 1 - \chi k^{-1/2}$$

and

$$\|(1 - \chi k^{-1/2})^{-1} X_+^{1/2} S'_+ X_+^{1/2} - I\|_2 \leq \kappa.$$

Thus, the general situation indeed can be reduced to the one with $\mu = 1$, $\mu_+ = 1 - \chi k^{-1/2}$, and we loose nothing assuming, in addition to what was already postulated, that

$$\mu \equiv t^{-1} \equiv \frac{\text{Tr}(XS)}{k} = 1, \quad \mu_+ = 1 - \chi k^{-1/2},$$

whence

$$[\text{Tr}(XS) =] \quad \sum_{i=1}^k x_i s_i = k \quad (4.5.30)$$

and

$$[\|tX^{1/2}SX^{1/2} - I\|_2^2 \equiv] \quad \sum_{i=1}^n (x_i s_i - 1)^2 \leq \kappa^2. \quad (4.5.31)$$

³⁰. We start with proving that (4.5.27) indeed has a unique solution. It is convenient to pass in (4.5.27) from the unknowns ΔX , ΔS to the unknowns

$$\begin{aligned} \delta X &= X^{-1/2} \Delta X X^{-1/2} &\Leftrightarrow \Delta X &= X^{1/2} \delta X X^{1/2}, \\ \delta S &= X^{1/2} \Delta S X^{1/2} &\Leftrightarrow \Delta S &= X^{-1/2} \delta S X^{-1/2}. \end{aligned} \quad (4.5.32)$$

With respect to the new unknowns, (4.5.27) becomes

$$\begin{aligned} (a) \quad & X^{1/2} \delta X X^{1/2} \in \mathcal{L}, \\ (b) \quad & X^{-1/2} \delta S X^{-1/2} \in \mathcal{L}^\perp, \\ (c) \quad & X^{1/2} \delta X X^{1/2} S + X^{1/2} \delta S X^{-1/2} + X^{-1/2} \delta S X^{1/2} + S X^{1/2} \delta X X^{1/2} = 2\mu_+ I - 2XS \\ (d) \quad L(\delta X, \delta S) \equiv & \left[\underbrace{\sqrt{x_i x_j} (s_i + s_j)}_{\phi_{ij}} (\delta X)_{ij} + \left(\underbrace{\sqrt{\frac{x_i}{x_j}} + \sqrt{\frac{x_j}{x_i}}}_{\psi_{ij}} \right) (\delta S)_{ij} \right]_{i,j=1}^k = 2[(\mu_+ - x_i s_i) \delta_{ij}]_{i,j=1}^k, \end{aligned} \quad (4.5.33)$$

where $\delta_{ij} = \begin{cases} 0, & i \neq j \\ 1, & i = j \end{cases}$ are the Kronecker symbols.

We first claim that (4.5.33), regarded as a system with unknown *symmetric* matrices δX , δS has a unique solution. Observe that (4.5.33) is a system with $2\dim \mathbf{S}^k \equiv 2N$ scalar unknowns and $2N$ scalar linear equations. Indeed, (4.5.33.a) is a system of $N' \equiv N - \dim \mathcal{L}$ linear equations, (4.5.33.b) is a system of $N'' = N - \dim \mathcal{L}^\perp = \dim \mathcal{L}$ linear equations, and (4.5.33.c) has N equations, so that the total # of linear equations in our system is $N' + N'' + N = (N - \dim \mathcal{L}) + \dim \mathcal{L} + N = 2N$. Now, to verify that the *square* system of linear equations (4.5.33) has exactly one solution, it suffices to prove that the homogeneous system

$$X^{1/2} \delta X X^{1/2} \in \mathcal{L}, \quad X^{-1/2} \delta S X^{-1/2} \in \mathcal{L}^\perp, \quad L(\delta X, \delta S) = 0$$

has only trivial solution. Let $(\delta X, \delta S)$ be a solution to the homogeneous system. Relation $L(\delta X, \Delta S) = 0$ means that

$$(\delta X)_{ij} = -\frac{\psi_{ij}}{\phi_{ij}} (\delta S)_{ij}, \quad (4.5.34)$$

whence

$$\mathrm{Tr}(\delta X \delta S) = - \sum_{i,j} \frac{\psi_{ij}}{\phi_{ij}} (\Delta S)_{ij}^2. \quad (4.5.35)$$

Representing $\delta X, \delta S$ via $\Delta X, \Delta S$ according to (4.5.32), we get

$$\mathrm{Tr}(\delta X \delta S) = \mathrm{Tr}(X^{-1/2} \Delta X X^{-1/2} X^{1/2} \Delta S X^{1/2}) = \mathrm{Tr}(X^{-1/2} \Delta X \Delta S X^{1/2}) = \mathrm{Tr}(\Delta X \Delta S),$$

and the latter quantity is 0 due to $\Delta X = X^{1/2} \delta X X^{1/2} \in \mathcal{L}$ and $\Delta S = X^{-1/2} \delta S X^{-1/2} \in \mathcal{L}^\perp$. Thus, the left hand side in (4.5.35) is 0; since $\phi_{ij} > 0, \psi_{ij} > 0$, (4.5.35) implies that $\delta S = 0$. But then $\delta X = 0$ in view of (4.5.34). Thus, the homogeneous version of (4.5.33) has the trivial solution only, so that (4.5.33) is solvable with a unique solution.

4^0 . Let $\delta X, \delta S$ be the unique solution to (4.5.33), and let $\Delta X, \Delta S$ be linked to $\delta X, \delta S$ according to (4.5.32). Our local goal is to bound from above the Frobenius norms of δX and δS .

From (4.5.33.c) it follows (cf. derivation of (4.5.35)) that

$$\begin{aligned} (a) \quad (\delta X)_{ij} &= -\frac{\psi_{ij}}{\phi_{ij}} (\delta S)_{ij} + 2 \frac{\mu_+ - x_i s_i}{\phi_{ii}} \delta_{ij}, \quad i, j = 1, \dots, k; \\ (b) \quad (\delta S)_{ij} &= -\frac{\phi_{ij}}{\psi_{ij}} (\delta X)_{ij} + 2 \frac{\mu_+ - x_i s_i}{\psi_{ii}} \delta_{ij}, \quad i, j = 1, \dots, k. \end{aligned} \quad (4.5.36)$$

Same as in the concluding part of 3^0 , relations (4.5.33.a – b) imply that

$$\mathrm{Tr}(\Delta X \Delta S) = \mathrm{Tr}(\delta X \delta S) = \sum_{i,j} (\delta X)_{ij} (\delta S)_{ij} = 0. \quad (4.5.37)$$

Multiplying (4.5.36.a) by $(\delta S)_{ij}$ and taking sum over i, j , we get, in view of (4.5.37), the relation

$$\sum_{i,j} \frac{\psi_{ij}}{\phi_{ij}} (\delta S)_{ij}^2 = 2 \sum_i \frac{\mu_+ - x_i s_i}{\phi_{ii}} (\delta S)_{ii}; \quad (4.5.38)$$

by “symmetric” reasoning, we get

$$\sum_{i,j} \frac{\phi_{ij}}{\psi_{ij}} (\delta X)_{ij}^2 = 2 \sum_i \frac{\mu_+ - x_i s_i}{\psi_{ii}} (\delta X)_{ii}. \quad (4.5.39)$$

Now let

$$\theta_i = x_i s_i, \quad (4.5.40)$$

so that in view of (4.5.30) and (4.5.31) one has

$$\begin{aligned} (a) \quad \sum_i \theta_i &= k, \\ (b) \quad \sum_i (\theta_i - 1)^2 &\leq \kappa^2. \end{aligned} \quad (4.5.41)$$

Observe that

$$\phi_{ij} = \sqrt{x_i x_j} (s_i + s_j) = \sqrt{x_i x_j} \left(\frac{\theta_i}{x_i} + \frac{\theta_j}{x_j} \right) = \theta_j \sqrt{\frac{x_i}{x_j}} + \theta_i \sqrt{\frac{x_j}{x_i}}.$$

Thus,

$$\begin{aligned} \phi_{ij} &= \theta_j \sqrt{\frac{x_i}{x_j}} + \theta_i \sqrt{\frac{x_j}{x_i}}, \\ \psi_{ij} &= \sqrt{\frac{x_i}{x_j}} + \sqrt{\frac{x_j}{x_i}}; \end{aligned} \quad (4.5.42)$$

since $1 - \kappa \leq \theta_i \leq 1 + \kappa$ by (4.5.41.b), we get

$$1 - \kappa \leq \frac{\phi_{ij}}{\psi_{ij}} \leq 1 + \kappa. \quad (4.5.43)$$

By the geometric-arithmetic mean inequality we have $\psi_{ij} \geq 2$, whence in view of (4.5.43)

$$\phi_{ij} \geq (1 - \kappa)\psi_{ij} \geq 2(1 - \kappa) \quad \forall i, j. \quad (4.5.44)$$

We now have

$$\begin{aligned} (1 - \kappa) \sum_{i,j} (\delta X)_{ij}^2 &\leq \sum_{i,j} \frac{\phi_{ij}}{\psi_{ij}} (\delta X)_{ij}^2 && [\text{see (4.5.43)}] \\ &\leq 2 \sum_i \frac{\mu_+ - x_i s_i}{\phi_{ii}} (\delta X)_{ii} && [\text{see (4.5.39)}] \\ &\leq 2 \sqrt{\sum_i (\mu_+ - x_i s_i)^2} \sqrt{\sum_i \psi_{ij}^{-2} (\delta X)_{ii}^2} \\ &\leq \sqrt{\sum_i ((1 - \theta_i)^2 - 2\chi k^{-1/2}(1 - \theta_i) + \chi^2 k^{-1})} \sqrt{\sum_{i,j} (\delta X)_{ij}^2} && [\text{see (4.5.44)}] \\ &\leq \sqrt{\chi^2 + \sum_i (1 - \theta_i)^2} \sqrt{\sum_{i,j} (\delta X)_{ij}^2} && [\text{since } \sum_i (1 - \theta_i) = 0 \text{ by (4.5.41.a)}] \\ &\leq \sqrt{\chi^2 + \kappa^2} \sqrt{\sum_{i,j} (\delta X)_{ij}^2} && [\text{see (4.5.41.b)}] \end{aligned}$$

and from the resulting inequality it follows that

$$\|\delta X\|_2 \leq \rho = \frac{\sqrt{\chi^2 + \kappa^2}}{1 - \kappa}. \quad (4.5.45)$$

Similarly,

$$\begin{aligned} (1 + \kappa)^{-1} \sum_j (\delta S)_{ij}^2 &\leq \sum_{i,j} \frac{\psi_{ij}}{\phi_{ij}} (\delta S)_{ij}^2 && [\text{see (4.5.43)}] \\ &\leq 2 \sum_i \frac{\mu_+ - x_i s_i}{\phi_{ii}} (\delta S)_{ii} && [\text{see (4.5.38)}] \\ &\leq 2 \sqrt{\sum_i (\mu_+ - x_i s_i)^2} \sqrt{\sum_i \phi_{ij}^{-2} (\delta S)_{ii}^2} \\ &\leq (1 - \kappa)^{-1} \sqrt{\sum_i (\mu_+ - \theta_i)^2} \sqrt{\sum_{i,j} (\delta S)_{ij}^2} && [\text{see (4.5.44)}] \\ &\leq (1 - \kappa)^{-1} \sqrt{\chi^2 + \kappa^2} \sqrt{\sum_{i,j} (\delta S)_{ij}^2} && [\text{same as above}] \end{aligned}$$

and from the resulting inequality it follows that

$$\|\delta S\|_2 \leq \frac{(1 + \kappa) \sqrt{\chi^2 + \kappa^2}}{1 - \kappa} = (1 + \kappa) \rho. \quad (4.5.46)$$

5⁰. We are ready to prove 2⁰.(i-ii). We have

$$X_+ = X + \Delta X = X^{1/2} (I + \delta X) X^{1/2},$$

and the matrix $I + \delta X$ is positive definite due to (4.5.45) (indeed, the right hand side in (4.5.45) is $\rho \leq 1$, whence the Frobenius norm (and therefore - the maximum of modulae of eigenvalues) of δX is less than 1). Note that by the just indicated reasons $I + \delta X \preceq (1 + \rho)I$, whence

$$X_+ \preceq (1 + \rho)X. \quad (4.5.47)$$

Similarly, the matrix

$$S_+ = S + \Delta S = X^{-1/2}(X^{1/2}SX^{1/2} + \delta S)X^{-1/2}$$

is positive definite. Indeed, the eigenvalues of the matrix $X^{1/2}SX^{1/2}$ are $\geq \min_i \theta_i \geq 1 - \kappa$, while the modulae of eigenvalues of δS , by (4.5.46), do not exceed $\frac{(1+\kappa)\sqrt{\chi^2+\kappa^2}}{1-\kappa} < 1 - \kappa$. Thus, the matrix $X^{1/2}SX^{1/2} + \delta S$ is positive definite, whence S_+ also is so. We have proved 2⁰(i).

2⁰.(ii) is easy to verify. First, by (4.5.33), we have $\Delta X \in \mathcal{L}$, $\Delta S \in \mathcal{L}^\perp$, and since $X \in \mathcal{L} - B$, $S \in \mathcal{L}^\perp + C$, we have $X_+ \in \mathcal{L} - B$, $S_+ \in \mathcal{L}^\perp + C$. Second, we have

$$\begin{aligned} \text{Tr}(X_+S_+) &= \text{Tr}(XS + X\Delta S + \Delta XS + \Delta X\Delta S) \\ &= \text{Tr}(XS + X\Delta S + \Delta XS) \\ &\quad [\text{since } \text{Tr}(\Delta X\Delta S) = 0 \text{ due to } \Delta X \in \mathcal{L}, \Delta S \in \mathcal{L}^\perp] \\ &= \mu_+ k \\ &\quad [\text{take the trace of both sides in (4.5.27.c)}] \end{aligned}$$

2⁰.(ii) is proved.

6⁰. It remains to verify 2⁰.(iii). We should bound from above the quantity

$$\Omega = \|\mu_+^{-1}X_+^{1/2}S_+X_+^{1/2} - I\|_2 = \|X_+^{1/2}(\mu_+^{-1}S_+ - X_+^{-1})X_+^{1/2}\|_2,$$

and our plan is first to bound from above the “close” quantity

$$\begin{aligned} \widehat{\Omega} &= \|X^{1/2}(\mu_+^{-1}S_+ - X_+^{-1})X^{1/2}\|_2 = \mu_+^{-1}\|Z\|_2, \\ Z &= X^{1/2}(S_+ - \mu_+X_+^{-1})X^{1/2}, \end{aligned} \tag{4.5.48}$$

and then to bound Ω in terms of $\widehat{\Omega}$.

6⁰.1. *Bounding $\widehat{\Omega}$* . We have

$$\begin{aligned} Z &= X^{1/2}(S_+ - \mu_+X_+^{-1})X^{1/2} \\ &= X^{1/2}(S + \Delta S)X^{1/2} - \mu_+X^{1/2}[X + \Delta X]^{-1}X^{1/2} \\ &= XS + \delta S - \mu_+X^{1/2}[X^{1/2}(I + \delta X)X^{1/2}]^{-1}X^{1/2} \\ &\quad [\text{see (4.5.32)}] \\ &= XS + \delta S - \mu_+(I + \delta X)^{-1} \\ &= XS + \delta S - \mu_+(I - \delta X) - \mu_+[(I + \delta X)^{-1} - I + \delta X] \\ &= \underbrace{XS + \delta S + \delta X - \mu_+I}_{Z^1} + \underbrace{(\mu_+ - 1)\delta X}_{Z^2} + \underbrace{\mu_+[I - \delta X - (I + \delta X)^{-1}]}_{Z^3}, \end{aligned}$$

so that

$$\|Z\|_2 \leq \|Z^1\|_2 + \|Z^2\|_2 + \|Z^3\|_2. \tag{4.5.49}$$

We are about to bound separately all 3 terms in the right hand side of the latter inequality.

Bounding $\|Z^2\|_2$: We have

$$\|Z^2\|_2 = |\mu_+ - 1|\|\delta X\|_2 \leq \chi k^{-1/2} \rho \tag{4.5.50}$$

(see (4.5.45) and take into account that $\mu_+ - 1 = -\chi k^{-1/2}$).

Bounding $\|Z^3\|_2$: Let λ_i be the eigenvalues of δX . We have

$$\begin{aligned}
 \|Z^3\|_2 &= \|\mu_+[(I + \delta X)^{-1} - I + \delta X]\|_2 \\
 &\leq \|(I + \delta X)^{-1} - I + \delta X\|_2 \\
 &= \sqrt{\sum_i \left(\frac{1}{1+\lambda_i} - 1 + \lambda_i\right)^2} \\
 &\quad [\text{since } |\mu_+| \leq 1] \\
 &= \sqrt{\sum_i \frac{\lambda_i^4}{(1+\lambda_i)^2}} \\
 &\leq \sqrt{\sum_i \frac{\rho^2 \lambda_i^2}{(1-\rho)^2}} \\
 &\quad [\text{see (4.5.45) and note that } \sum_i \lambda_i^2 = \|\delta X\|_2^2 \leq \rho^2] \\
 &\leq \frac{\rho^2}{1-\rho}
 \end{aligned} \tag{4.5.51}$$

Bounding $\|Z^1\|_2$: This is a bit more involving. We have

$$\begin{aligned}
 Z_{ij}^1 &= (XS)_{ij} + (\delta S)_{ij} + (\delta X)_{ij} - \mu_+ \delta_{ij} \\
 &= (\delta X)_{ij} + (\delta S)_{ij} + (x_i s_i - \mu_+) \delta_{ij} \\
 &= (\delta X)_{ij} \left[1 - \frac{\phi_{ij}}{\psi_{ij}} \right] + \left[2 \frac{\mu_+ - x_i s_i}{\psi_{ii}} + x_i s_i - \mu_+ \right] \delta_{ij} \\
 &\quad [\text{we have used (4.5.36.b)}] \\
 &= (\delta X)_{ij} \left[1 - \frac{\phi_{ij}}{\psi_{ij}} \right] \\
 &\quad [\text{since } \psi_{ii} = 2, \text{ see (4.5.42)}]
 \end{aligned}$$

whence, in view of (4.5.43),

$$|Z_{ij}^1| \leq \left| 1 - \frac{1}{1-\kappa} \right| |(\delta X)_{ij}| = \frac{\kappa}{1-\kappa} |(\delta X)_{ij}|,$$

so that

$$\|Z^1\|_2 \leq \frac{\kappa}{1-\kappa} \|\delta X\|_2 \leq \frac{\kappa}{1-\kappa} \rho \tag{4.5.52}$$

(the concluding inequality is given by (4.5.45)).

Assembling (4.5.50), (4.5.51), (4.5.52) and (4.5.49), we come to

$$\|Z\|_2 \leq \rho \left[\frac{\chi}{\sqrt{k}} + \frac{\rho}{1-\rho} + \frac{\kappa}{1-\kappa} \right],$$

whence, by (4.5.48),

$$\widehat{\Omega} \leq \frac{\rho}{1-\chi k^{-1/2}} \left[\frac{\chi}{\sqrt{k}} + \frac{\rho}{1-\rho} + \frac{\kappa}{1-\kappa} \right]. \tag{4.5.53}$$

$6^0.2.$ Bounding Ω . We have

$$\begin{aligned}
\Omega^2 &= \|\mu_+^{-1} X_+^{1/2} S_+ X_+^{1/2} - I\|_2^2 \\
&= \|X_+^{1/2} \underbrace{[\mu_+^{-1} S_+ - X_+^{-1}]}_{\Theta = \Theta^T} X_+^{1/2}\|_2^2 \\
&= \text{Tr} \left(X_+^{1/2} \Theta X_+ \Theta X_+^{1/2} \right) \\
&\leq (1 + \rho) \text{Tr} \left(X_+^{1/2} \Theta X \Theta X_+^{1/2} \right) \quad [\text{see (4.5.47)}] \\
&= (1 + \rho) \text{Tr} \left(X_+^{1/2} \Theta X^{1/2} X^{1/2} \Theta X_+^{1/2} \right) \\
&= (1 + \rho) \text{Tr} \left(X^{1/2} \Theta X_+^{1/2} X_+^{1/2} \Theta X^{1/2} \right) \\
&= (1 + \rho) \text{Tr} \left(X^{1/2} \Theta X_+ \Theta X^{1/2} \right) \\
&\leq (1 + \rho)^2 \text{Tr} \left(X^{1/2} \Theta X \Theta X^{1/2} \right) \quad [\text{the same (4.5.47)}] \\
&= (1 + \rho)^2 \|X^{1/2} \Theta X^{1/2}\|_2^2 \\
&= (1 + \rho)^2 \|X^{1/2} [\mu_+^{-1} S_+ - X_+^{-1}] X^{1/2}\|_2^2 \\
&= (1 + \rho)^2 \widehat{\Omega}^2 \quad [\text{see (4.5.48)}]
\end{aligned}$$

so that

$$\begin{aligned}
\Omega &\leq (1 + \rho) \widehat{\Omega} = \frac{\rho(1+\rho)}{1-\chi k^{-1/2}} \left[\frac{\chi}{\sqrt{k}} + \frac{\rho}{1-\rho} + \frac{\kappa}{1-\kappa} \right], \\
\rho &= \frac{\sqrt{\chi^2 + \kappa^2}}{1-\kappa}.
\end{aligned} \tag{4.5.54}$$

(see (4.5.53) and (4.5.45)).

It is immediately seen that if $0 < \chi \leq \kappa \leq 0.1$, the right hand side in the resulting bound for Ω is $\leq \kappa$, as required in $2^0.\text{(iii)}$. ■

Remark 4.5.2 We have carried out the complexity analysis for a large group of primal-dual path-following methods for SDP (i.e., for the case of $\mathbf{K} = \mathbf{S}_+^k$). In fact, the constructions and the analysis we have presented can be word by word extended to the case when \mathbf{K} is a direct product of semidefinite cones – you just should bear in mind that all symmetric matrices we deal with, like the primal and the dual solutions X, S , the scaling matrices Q , the primal-dual search directions $\Delta X, \Delta S$, etc., are block-diagonal with common block-diagonal structure. In particular, our constructions and analysis work for the case of LP – this is the case when \mathbf{K} is a direct product of one-dimensional semidefinite cones. Note that in the case of LP Zhang's family of primal-dual search directions reduces to a single direction: since now X, S, Q are diagonal matrices, the scaling (4.5.17) \mapsto (4.5.18) does not vary the equations of augmented complementary slackness.

The recipe to translate all we have presented for the case of SDP to the case of LP is very simple: in the above text, you should assume all matrices like X, S, \dots to be diagonal and look what the operations with these matrices required by the description of the method do with their diagonals. By the way, one of the very first approaches to the design and the analysis of IP methods for SDP was exactly opposite: you take an IP scheme for LP, replace in its description the words “nonnegative vectors” with “positive semidefinite diagonal matrices” and then erase the adjective “diagonal”.

4.6 Complexity bounds for LP, CQP, SDP

In what follows we list the best known so far complexity bounds for LP, CQP and SDP. These bounds are yielded by IP methods and, essentially, say that the *Newton complexity* of finding ϵ -solution to an instance – the total # of steps of a “good” IP algorithm before an ϵ -solution is found – is $O(1)\sqrt{\theta(K)} \ln \frac{1}{\epsilon}$. This is what should be expected in view of discussion in Section 4.5.2; note, however, that the complexity bounds to follow take into account the necessity to “reach the highway” – to come close to the central path before tracing it, while in Section 4.5.2 we were focusing on how fast could we reduce the duality gap after the central path (“the highway”) is reached.

Along with complexity bounds expressed in terms of the Newton complexity, we present the bounds on the number of operations of Real Arithmetic required to build an ϵ -solution. Note that these latter bounds typically are conservative – when deriving them, we assume the data of an instance “completely unstructured”, which is usually not the case (cf. Warning in Section 4.5.2); exploiting structure of the data, one usually can reduce significantly computational effort per step of an IP method and consequently – the arithmetic cost of ϵ -solution.

4.6.1 Complexity of \mathcal{LP}_b

Family of problems:

Problem instance: a program

$$\min_x \left\{ c^T x : a_i^T x \leq b_i, i = 1, \dots, m; \|x\|_2 \leq R \right\} \quad [x \in \mathbf{R}^n]; \quad (p)$$

Data:

$$\text{Data}(p) = [m; n; c; a_1, b_1; \dots; a_m, b_m; R],$$

$$\text{Size}(p) \equiv \dim \text{Data}(p) = (m+1)(n+1) + 2.$$

ϵ -solution: an $x \in \mathbf{R}^n$ such that

$$\begin{aligned} \|x\|_\infty &\leq R, \\ a_i^T x &\leq b_i + \epsilon, \quad i = 1, \dots, m, \\ c^T x &\leq \text{Opt}(p) + \epsilon \end{aligned}$$

(as always, the optimal value of an infeasible problem is $+\infty$).

Newton complexity of ϵ -solution: ¹⁴⁾

$$\text{Compl}^{\text{Nwt}}(p, \epsilon) = O(1)\sqrt{m+n}\text{Digits}(p, \epsilon),$$

where

$$\text{Digits}(p, \epsilon) = \ln \left(\frac{\text{Size}(p) + \|\text{Data}(p)\|_1 + \epsilon^2}{\epsilon} \right)$$

is the number of accuracy digits in ϵ -solution, see Section 4.1.2.

¹⁴⁾In what follows, the precise meaning of a statement “the Newton/arithmetic complexity of finding ϵ -solution of an instance (p) does not exceed N ” is as follows: as applied to the input $(\text{Data}(p), \epsilon)$, the method underlying our bound terminates in no more than N steps (respectively, N arithmetic operations) and outputs either a vector, which is an ϵ -solution to the instance, or the correct conclusion “ (p) is infeasible”.

Arithmetic complexity of ϵ -solution:

$$\text{Compl}(p, \epsilon) = O(1)(m+n)^{3/2}n^2\text{Digits}(p, \epsilon).$$

4.6.2 Complexity of \mathcal{CQP}_b

Family of problems:

Problem instance: a program

$$\min_x \left\{ c^T x : \|A_i x + b_i\|_2 \leq c_i^T x + d_i, \quad i = 1, \dots, m; \|x\|_2 \leq R \right\} \quad \begin{bmatrix} x \in \mathbf{R}^n \\ b_i \in \mathbf{R}^{k_i} \end{bmatrix} \quad (p)$$

Data:

$$\begin{aligned} \text{Data}(P) &= [m; n; k_1, \dots, k_m; c; A_1, b_1, c_1, d_1; \dots; A_m, b_m, c_m, d_m; R], \\ \text{Size}(p) &\equiv \dim \text{Data}(p) = (m + \sum_{i=1}^m k_i)(n+1) + m + n + 3. \end{aligned}$$

ϵ -solution: an $x \in \mathbf{R}^n$ such that

$$\begin{aligned} \|x\|_2 &\leq R, \\ \|A_i x + b_i\|_2 &\leq c_i^T x + d_i + \epsilon, \quad i = 1, \dots, m, \\ c^T x &\leq \text{Opt}(p) + \epsilon. \end{aligned}$$

Newton complexity of ϵ -solution:

$$\text{Compl}^{\text{Nwt}}(p, \epsilon) = O(1)\sqrt{m+1}\text{Digits}(p, \epsilon).$$

Arithmetic complexity of ϵ -solution:

$$\text{Compl}(p, \epsilon) = O(1)(m+1)^{1/2}n(n^2 + m + \sum_{i=0}^m k_i^2)\text{Digits}(p, \epsilon).$$

4.6.3 Complexity of \mathcal{SDP}_b

Family of problems:

Problem instance: a program

$$\min_x \left\{ c^T x : A_0 + \sum_{j=1}^n x_j A_j \succeq 0, \|x\|_2 \leq R \right\} \quad [x \in \mathbf{R}^n], \quad (p)$$

where A_j , $j = 0, 1, \dots, n$, are symmetric block-diagonal matrices with m diagonal blocks $A_j^{(i)}$ of sizes $k_i \times k_i$, $i = 1, \dots, m$.

Data:

$$\begin{aligned} \text{Data}(p) &= [m; n; k_1, \dots, k_m; c; A_0^{(1)}, \dots, A_0^{(m)}; \dots; A_n^{(1)}, \dots, A_n^{(m)}; R], \\ \text{Size}(p) &\equiv \dim \text{Data}(P) = \left(\sum_{i=1}^m \frac{k_i(k_i+1)}{2} \right) (n+1) + m + n + 3. \end{aligned}$$

ϵ -solution: an x such that

$$\begin{aligned}\|x\|_2 &\leq R, \\ A_0 + \sum_{j=1}^n x_j A_j &\succeq -\epsilon I, \\ c^T x &\leq \text{Opt}(p) + \epsilon.\end{aligned}$$

Newton complexity of ϵ -solution:

$$\text{Compl}^{\text{Nwt}}(p, \epsilon) = O(1)(1 + \sum_{i=1}^m k_i)^{1/2} \text{Digits}(p, \epsilon).$$

Arithmetic complexity of ϵ -solution:

$$\text{Compl}(p, \epsilon) = O(1)(1 + \sum_{i=1}^m k_i)^{1/2} n(n^2 + n \sum_{i=1}^m k_i^2 + \sum_{i=1}^m k_i^3) \text{Digits}(p, \epsilon).$$

4.7 Concluding remarks

We have discussed IP methods for LP, CQP and SDP as “mathematical animals”, with emphasis on the ideas underlying the algorithms and on the theoretical complexity bounds ensured by the methods. Now it is time to say a couple of words on software implementations of IP algorithms and on practical performance of the resulting codes.

As far as the performance of recent IP software is concerned, the situation heavily depends on whether we are speaking about codes for LP, or those for CQP and SDP.

- There exists extremely powerful commercial IP software for LP, capable to handle reliably really large-scale LP's and quite competitive with the best Simplex-type codes for Linear Programming. E.g., one of the best modern LP solvers – CPLEX – allows user to choose between a Simplex-type and IP modes of execution, and in many cases the second option reduces the running time by orders of magnitudes. With a state-of-the-art computer, CPLEX is capable to solve routinely real-world LP's with tens and hundreds thousands of variables and constraints; in the case of favourable structured constraint matrices, the numbers of variables and constraints can become as large as few millions.

- There already exists a very powerful commercial software for CQP – MOSEK (Erling Andersen, <http://www.mosek.com>). I would say that as far as LP (and even mixed integer programming) are concerned, MOSEK compares favourable to CPLEX, and it allows to solve really large CQP's of favourable structure.

- For the time being, IP software for SDP's is not as well-polished, reliable and powerful as the LP one. I would say that the codes available for the moment are capable to solve SDP's with no more than 1,000 – 1,500 design variables.

There are two groups of reasons making the power of SDP software available for the moment that inferior as compared to the capabilities of interior point LP and CQP solvers – the “historical” and the “intrinsic” ones. The “historical” aspect is simple: the development of IP software for LP, on one hand, and for SDP, on the other, has started, respectively, in the mid-eighties and the mid-nineties; for the time being (2002), this is definitely a difference. Well, being too young is the only shortcoming which for sure passes away... Unfortunately, there are intrinsic problems with IP algorithms for large-scale (many thousands of variables) SDP's. Recall that

the influence of the size of an SDP/CQP program on the complexity of its solving by an IP method is twofold:

– first, the size affects the Newton complexity of the process. Theoretically, the number of steps required to reduce the duality gap by a constant factor, say, factor 2, is proportional to $\sqrt{\theta(K)}$ ($\theta(K)$ is twice the total # of conic quadratic inequalities for CQP and the total row size of LMI's for SDP). Thus, we could expect an unpleasant growth of the iteration count with $\theta(K)$. Fortunately, the iteration count for good IP methods usually is much less than the one given by the worst-case complexity analysis and is typically about few tens, independently of $\theta(K)$.

– second, the larger is the instance, the larger is the system of linear equations one should solve to generate new primal (or primal-dual) search direction, and, consequently, the larger is the computational effort per step (this effort is dominated by the necessity to assemble and to solve the linear system). Now, the system to be solved depends, of course, on what is the IP method we are speaking about, but it newer is simpler (and for most of the methods, is not more complicated as well) than the system (4.5.8) arising in the primal path-following method:

$$\underbrace{\mathcal{A}^*[\nabla^2 K(\bar{X})]\mathcal{A}}_{\mathcal{H}} \Delta x = -\underbrace{[t_+ c + \mathcal{A}^* \nabla K(\bar{X})]}_h. \quad (\text{Nwt})$$

The size n of this system is exactly the design dimension of problem (CP).

In order to process (Nwt), one should assemble the system (compute \mathcal{H} and h) and then solve it. Whatever is the cost of assembling (Nwt), you should be able to store the resulting matrix \mathcal{H} in memory and to factorize the matrix in order to get the solution. *Both these problems – storing and factorizing \mathcal{H} – become prohibitively expensive when \mathcal{H} is a large dense¹⁵⁾ matrix.* (Think how happy you will be with the necessity to store $\frac{5000 \times 5001}{2} = 12,502,500$ reals representing a dense 5000×5000 symmetric matrix \mathcal{H} and with the necessity to perform $\approx \frac{5000^3}{6} \approx 2.08 \times 10^{10}$ arithmetic operations to find its Choleski factor).

The necessity to assemble and to solve large-scale systems of linear equations is intrinsic for IP methods as applied to large-scale optimization programs, and in this respect there is no difference between LP and CQP, on one hand, and SDP, on the other hand. The difference is in how difficult is to handle these large-scale linear systems. In real life LP's-CQP's-SDP's, the structure of the data allows to assemble (Nwt) at a cost negligibly small as compared to the cost of factorizing \mathcal{H} , which is a good news. Another good news is that in typical real world LP's, and to some extent for real-world CQP's, \mathcal{H} turns out to be “very well-structured”, which reduces dramatically the expenses required by factorizing the matrix and storing the Choleski factor. All practical IP solvers for LP and CQP utilize these favourable properties of real life problems, and this is where their ability to solve problems with tens/hundreds thousands of variables and constraints comes from. Spoil the structure of the problem – and an IP method will be unable to solve an LP with just few thousands of variables. Now, in contrast to real life LP's and CQP's, real life SDP's typically result in dense matrices \mathcal{H} , and this is where severe limitations on the sizes of “tractable in practice” SDP's come from. In this respect, real life CQP's are somewhere in-between LP's and SDP's, so that the sizes of “tractable in practice” CQP's could be significantly larger than in the case of SDP's.

It should be mentioned that assembling matrices of the linear systems we are interested in and solving these systems by the standard Linear Algebra techniques is not the only possible way to implement an IP method. Another option is to solve these linear systems by iterative

¹⁵⁾I.e., with $O(n^2)$ nonzero entries.

methods. With this approach, all we need to solve a system like (Nwt) is a possibility to multiply a given vector by the matrix of the system, and this does *not* require assembling and storing in memory the matrix itself. E.g., to multiply a vector Δx by \mathcal{H} , we can use the multiplicative representation of \mathcal{H} as presented in (Nwt). *Theoretically*, the outlined iterative schemes, as applied to real life SDP's, allow to reduce by orders of magnitudes the arithmetic cost of building search directions and to avoid the necessity to assemble and store huge dense matrices, which is an extremely attractive opportunity. The difficulty, however, is that the iterative schemes are much more affected by rounding errors than the usual Linear Algebra techniques; as a result, *for the time being* “iterative-Linear-Algebra-based” implementation of IP methods is no more than a challenging goal.

Although the sizes of SDP's which can be solved with the existing codes are not that impressive as those of LP's, the possibilities offered to a practitioner by SDP IP methods could hardly be overestimated. Just ten years ago we could not even dream of solving an SDP with more than few tens of variables, while today we can solve routinely 20-25 times larger SDP's, and we have all reasons to believe in further significant progress in this direction.

Lecture 5

Simple methods for extremely large-scale problems

5.1 Motivation: Why simple methods?

The polynomial time Interior Point methods, same as all other polynomial time methods for Convex Programming known so far, have a not that pleasant common feature: the arithmetic cost C of an iteration in such a method grows *nonlinearly* with the design dimension n of the problem, unless the problem possesses a very favourable structure. E.g., in IP methods, an iteration requires solving a system of linear equations with (at least) n unknowns. To solve this auxiliary problem, it costs at least $O(n^2)$ operations (with the traditional Linear Algebra – even $O(n^3)$ operations), except for the cases when the matrix of the system is very sparse and, moreover, possesses a well-structured sparsity pattern. The latter indeed is the case when solving most of LPs of decision-making origin, but often is not the case for LPs coming from Engineering, and nearly never is the case for SDPs. For other known polynomial time methods, the situation is similar – the arithmetic cost of an iteration, even in the case of extremely simple objectives and feasible sets, is at least $O(n^2)$. With n of order of tens and hundreds of thousands, the computational effort of $O(n^2)$, not speaking about $O(n^3)$, operations per iteration becomes prohibitively large – basically, you never will finish the very first iteration of your method... On the other hand, the design dimensions of tens and hundreds of thousands is exactly what is met in many applications, like SDP relaxations of combinatorial problems involving large graphs or Structural Design (especially for 3D structures). As another important application of this type, consider 3D Medical Imaging problem arising in Positron Emission Tomography.

Positron Emission Tomography (PET) is a powerful, non-invasive, medical diagnostic imaging technique for measuring the metabolic activity of cells in the human body. It has been in clinical use since the early 1990s. PET imaging is unique in that it shows the *chemical functioning* of organs and tissues, while other imaging techniques - such as X-ray, computerized tomography (CT) and magnetic resonance imaging (MRI) - show *anatomic structures*.

A PET scan involves the use of a *radioactive tracer* – a fluid with a small amount of a radioactive material which has the property of emitting positrons. When the tracer is administered to a patient, either by injection or inhalation of gas, it distributes within the body. For a properly chosen tracer, this distribution “concentrates” in desired locations, e.g., in the areas of high metabolic activity where cancer tumors can be expected.

The radioactive component of the tracer disintegrates, emitting positrons. Such a positron nearly immediately annihilates with a near-by electron, giving rise to two γ -quants flying at the speed of light off the point of annihilation in nearly opposite directions along a line with a completely random orientation (i.e., line's direction is drawn at random from the uniform distribution on the unit sphere in 3D). The

γ -quants penetrate the surrounding tissue and are registered outside the patient by a PET scanner consisting of circular arrays (*rings*) of gamma radiation detectors. Since the two gamma rays are emitted simultaneously and travel in almost exactly opposite directions, we can say a lot on the location of their source: when a pair of detectors register high-energy γ -quants within a short ($\sim 10^{-8}$ sec) time window (“a coincidence event”), we know that the photons came from a disintegration act, and that the act took place on the line (“line of response” (LOR)) linking the detectors. The measured data set is the collection of numbers of coincidences counted by different pairs of detectors (“bins”), and the problem is to recover from these measurements the 3D density of the tracer.

The mathematical model of the process, after appropriate discretization, is

$$y = P\lambda + \xi,$$

where

- $\lambda \geq 0$ is the vector representing the (discretized) density of the tracer; the entries of λ are indexed by voxels – small cubes into which we partition the field of view, and λ_j is the mean density of the tracer in voxel j . Typically, the number n of voxels is in the range from 3×10^5 to 3×10^6 , depending on the resolution of the discretization grid;
- y are the measurements; the entries in y are indexed by bins – pairs of detectors, and y_i is the number of coincidences counted by i -th pair of detectors. Typically, the dimension m of y – the total number of bins – is millions (at least 3×10^6);
- P is the projection matrix; its entries p_{ij} are the probabilities for a LOR originating in voxel j to be registered by bin i . These probabilities are readily given by the geometry of the scanner;
- ξ is the measurement noise coming mainly from the fact that all physical processes underlying PET are random. The standard statistical model for PET implies that y_i , $i = 1, \dots, m$, are independent Poisson random variables with the expectations $(P\lambda)_i$.

The problem we are interested in is to recover tracer’s density λ given measurements y . As far as the quality of the result is concerned, the most attractive reconstruction scheme is given by the standard in Statistics *Likelihood Ratio* maximization: denoting $p(\cdot|\lambda)$ the density, taken w.r.t. an appropriate dominating measure, of the probability distribution of the measurements coming from λ , the estimate of the unknown true value λ_* of λ is

$$\hat{\lambda} = \underset{\lambda \geq 0}{\operatorname{argmin}} p(y|\lambda),$$

where y is the vector of measurements.

For the aforementioned Poisson model of PET, building the Maximum Likelihood estimate is equivalent to solving the optimization problem

$$\min_{\lambda} \left\{ \sum_{j=1}^n \lambda_j p_j - \sum_{i=1}^m y_i \ln \left(\sum_{j=1}^n \lambda_j p_{ij} \right) : \lambda \geq 0 \right\} \quad \left[p_j = \sum_i p_{ij} \right] \quad (\text{PET})$$

This is a nicely structured convex program (by the way, polynomially reducible to CQP and even LP). The only difficulty – and a severe one – is in huge sizes of the problem: as it was already explained, the number n of decision variables is at least 300,000, while the number m of log-terms in the objective is in the range from 3×10^6 to 25×10^6 .

At the present level of our knowledge, the design dimension n of order of tens and hundreds of thousands rules out the possibility to solve a *nonlinear* convex program, even a well-structured one, by polynomial time methods because of at least quadratic in n “blowing up” the arithmetic cost of an iteration. When n is really large, all we can use are simple methods with linear in n cost of an iteration. As a byproduct of this restriction, we *cannot utilize anymore our knowledge*

of the analytic structure of the problem, since all known for the time being ways of doing so are too expensive, provided that n is large. As a result, we are enforced to restrict ourselves with black-box-oriented optimization techniques – those which use solely the possibility to compute the values and the (sub)gradients of the objective and the constraints at a point. In Convex Optimization, two types of “cheap” black-box-oriented optimization techniques are known:

- techniques for *unconstrained* minimization of *smooth* convex functions (Gradient Descent, Conjugate Gradients, quasi-Newton methods with restricted memory, etc.);
- subgradient-type techniques, called also *First Order algorithms*, for *nonsmooth* convex programs, including constrained ones.

Since the majority of applications are constrained, we restrict our exposition to the techniques of the second type. We start with investigating of what, *in principle*, can be expected of black-box-oriented optimization techniques.

5.1.1 Black-box-oriented methods and Information-based complexity

Consider a Convex Programming program in the form

$$\min_x \{f(x) : x \in X\}, \quad (\text{CP})$$

where X is a convex compact set in a Euclidean space (which we w.l.o.g. identify with \mathbf{R}^n) and the objective f is a continuous convex function on \mathbf{R}^n . Let us fix a family $\mathcal{P}(X)$ of convex programs (CP) with X common for all programs from the family, so that such a program can be identified with the corresponding objective, and the family itself is nothing but certain family of convex functions on \mathbf{R}^n . We intend to explain what is the *Information-based complexity* of $\mathcal{P}(X)$ – informally, complexity of the family w.r.t. “black-box-oriented” methods. We start with defining such a method as a routine \mathcal{B} as follows:

1. When starting to solve (CP), \mathcal{B} is given an accuracy $\epsilon > 0$ to which the problem should be solved and knows that the problem belongs to a given family $\mathcal{P}(X)$. However, \mathcal{B} does *not* know what is the particular problem it deals with.
2. In course of solving the problem, \mathcal{B} has an access to the First Order oracle for f . This oracle is capable, given

on input a point $x \in \mathbf{R}^n$, to report on output what is the value $f(x)$ and a subgradient $f'(x)$ of f at x .

\mathcal{B} generates somehow a sequence of *search points* x_1, x_2, \dots and calls the First Order oracle to get the values and the subgradients of f at these points. The rules for building x_t can be arbitrary, *except for the fact that they should be non-anticipating (causal)*: x_t can depend only on the information $f(x_1), f'(x_1), \dots, f(x_{t-1}), f'(x_{t-1})$ on f accumulated by \mathcal{B} at the first $t - 1$ steps.

3. After certain number $T = T_{\mathcal{B}}(f, \epsilon)$ of calls to the oracle, \mathcal{B} terminates and outputs the result $z_{\mathcal{B}}(f, \epsilon)$. This result again should depend solely on the information on f accumulated by \mathcal{B} at the T search steps, and must be an ϵ -solution to (CP), i.e.,

$$z_{\mathcal{B}}(f, \epsilon) \in X \quad \& \quad f(z_{\mathcal{B}}(f, \epsilon)) - \min_X f \leq \epsilon.$$

We measure the complexity of $\mathcal{P}(X)$ w.r.t. a solution method \mathcal{B} by the function

$$\text{Compl}_{\mathcal{B}}(\epsilon) = \max_{f \in \mathcal{P}(X)} T_{\mathcal{B}}(f, \epsilon)$$

– by the minimal number of steps in which \mathcal{B} is capable to solve within accuracy ϵ every instance of $\mathcal{P}(X)$. Finally, the *Information-based complexity* of the family $\mathcal{P}(X)$ of problems is defined as

$$\text{Compl}(\epsilon) = \min_{\mathcal{B}} \text{Compl}_{\mathcal{B}}(\epsilon),$$

the minimum being taken over all solution methods. Thus, the relation $\text{Compl}(\epsilon) = N$ means, first, that there exists a solution method \mathcal{B} capable to solve within accuracy ϵ every instance of $\mathcal{P}(X)$ in no more than N calls to the First Order oracle, and, second, that for every solution method \mathcal{B} there exists an instance of $\mathcal{P}(X)$ such that \mathcal{B} solves the instance within the accuracy ϵ in at least N steps.

Note that as far as black-box-oriented optimization methods are concerned, the information-based complexity $\text{Compl}(\epsilon)$ of a family $\mathcal{P}(X)$ is a *lower bound* on “actual” computational effort, whatever it means, sufficient to find ϵ -solution to every instance of the family.

5.1.2 Main results on Information-based complexity of Convex Programming

Main results on Information-based complexity of Convex Programming can be summarized as follows. Let X be a solid in \mathbf{R}^n (a convex compact set with a nonempty interior), and let $\mathcal{P}(X)$ be the family of all convex functions on \mathbf{R}^n normalized by the condition

$$\max_X f - \min_X f \leq 1. \quad (5.1.1)$$

For this family,

- I. Complexity of finding high-accuracy solutions in fixed dimension is independent of the geometry of X . Specifically,

$$\begin{aligned} \forall (\epsilon \leq \epsilon(X)) : & O(1)n \ln\left(2 + \frac{1}{\epsilon}\right) \leq \text{Compl}(\epsilon); \\ \forall (\epsilon > 0) : & \text{Compl}(\epsilon) \leq O(1)n \ln\left(2 + \frac{1}{\epsilon}\right), \end{aligned} \quad (5.1.2)$$

where

- $O(1)$ ’s are appropriately chosen positive absolute constants,
- $\epsilon(X)$ depends on the geometry of X , but never is less than $\frac{1}{n^2}$, where n is the dimension of X .

- II. Complexity of finding solutions of fixed accuracy in high dimensions does depend on the geometry of X . Here are 3 typical results:

- (a) Let X be an n -dimensional box: $X = \{x \in \mathbf{R}^n : \|x\|_\infty \leq 1\}$. Then

$$\epsilon \leq \frac{1}{2} \Rightarrow O(1)n \ln\left(\frac{1}{\epsilon}\right) \leq \text{Compl}(\epsilon) \leq O(1)n \ln\left(\frac{1}{\epsilon}\right). \quad (5.1.3)$$

- (b) Let X be an n -dimensional ball: $X = \{x \in \mathbf{R}^n : \|x\|_2 \leq 1\}$. Then

$$n \geq \frac{1}{\epsilon^2} \Rightarrow \frac{O(1)}{\epsilon^2} \leq \text{Compl}(\epsilon) \leq \frac{O(1)}{\epsilon^2}. \quad (5.1.4)$$

(c) Let X be an n -dimensional hyperoctahedron: $X = \{x \in \mathbf{R}^n : \|x\|_1 \leq 1\}$. Then

$$n \geq \frac{1}{\epsilon^2} \Rightarrow \frac{O(1)}{\epsilon^2} \leq \text{Compl}(\epsilon) \leq \frac{O(\ln n)}{\epsilon^2} \quad (5.1.5)$$

(in fact, $O(1)$ in the lower bound can be replaced with $O(\ln n)$, provided that $n >> \frac{1}{\epsilon^2}$).

Since we are interested in extremely large-scale problems, the moral which we can extract from the outlined results is as follows:

- I is discouraging: it says that we have no hope to guarantee high accuracy, like $\epsilon = 10^{-6}$, when solving large-scale problems with black-box-oriented methods; indeed, with $O(n)$ steps per accuracy digit and at least $O(n)$ operations per step (this many operations are required already to input a search point to the oracle), the arithmetic cost per accuracy digit is at least $O(n^2)$, which is prohibitively large for really large n .

- II is partly discouraging, partly encouraging. A bad news reported by II is that when X is a box, which is the most typical situation in applications, we have no hope to solve extremely large-scale problems, in a reasonable time, to guaranteed, even low, accuracy, since the required number of steps should be at least of order of n . A good news reported by II is that there exist situations where the complexity of minimizing a convex function within a fixed accuracy is independent, or nearly independent, of the design dimension. Of course, the dependence of the complexity bounds in (5.1.4) and (5.1.5) on ϵ is very bad and has nothing in common with being polynomial in $\ln(1/\epsilon)$; however, this drawback is tolerable when we do not intend to get high accuracy. Another drawback is that there are not that many applications where the feasible set is a ball or a hyperoctahedron. Note, however, that in fact we can save the most important for us upper complexity bounds in (5.1.4) and (5.1.5) when requiring from X to be a subset of a ball, respectively, of a hyperoctahedron, rather than to be the entire ball/hyperoctahedron. This extension is not costless: we should simultaneously strengthen the normalization condition (5.1.1). Specifically, we shall see that

B. The upper complexity bound in (5.1.4) remains valid when $X \subset \{x : \|x\|_2 \leq 1\}$ and

$$\mathcal{P}(X) = \{f : f \text{ is convex and } |f(x) - f(y)| \leq \|x - y\|_2 \forall x, y \in X\};$$

S. The upper complexity bound in (5.1.5) remains valid when $X \subset \{x : \|x\|_1 \leq 1\}$ and

$$\mathcal{P}(X) = \{f : f \text{ is convex and } |f(x) - f(y)| \leq \|x - y\|_1 \forall x, y \in X\}.$$

Note that the “ball-like” case mentioned in B seems to be rather artificial: the Euclidean norm associated with this case is a very natural mathematical entity, but this is all we can say in its favour. For example, the normalization of the objective in B is that the Lipschitz constant of f w.r.t. $\|\cdot\|_2$ is ≤ 1 , or, which is the same, that the vector of the first order partial derivatives of f should, at every point, be of $\|\cdot\|_2$ -norm not exceeding 1. In other words, “typical” magnitudes of the partial derivatives of f should become smaller and smaller as the number of variables grows; what could be the reasons for such a strange behaviour? In contrast to this, the normalization condition imposed on f in S is that the Lipschitz constant of f w.r.t. $\|\cdot\|_1$ is ≤ 1 , or, which is the same, that the $\|\cdot\|_\infty$ -norm of the vector of partial derivatives of f is ≤ 1 . In other words, the normalization is that the magnitudes of the first order partial derivatives of f should be ≤ 1 , and this normalization is “dimension-independent”. Of course, in B we deal with minimization

over subsets of the unit ball, while in S we deal with minimization over the subsets of the unit hyperoctahedron, which is much smaller than the unit ball. However, there do exist problems in reality where we should minimize over the standard simplex

$$\Delta_n = \{x \in \mathbf{R}^n : x \geq 0, \sum_x x_i = 1\},$$

which indeed is a subset of the unit hyperoctahedron. For example, it turns out that the PET Image Reconstruction problem (PET) is in fact the problem of minimization over the standard simplex. Indeed, the optimality condition for (PET) reads

$$\lambda_j \left(p_j - \sum_i y_i \frac{p_{ij}}{\sum_\ell p_{i\ell} \lambda_\ell} \right) = 0, \quad j = 1, \dots, n;$$

summing up these equalities, we get

$$\sum_j p_j \lambda_j = B \equiv \sum_i y_i.$$

It follows that the optimal solution to (PET) remains unchanged when we add to the non-negativity constraints $\lambda_j \geq 0$ also the constraint $\sum_j p_j \lambda_j = B$. Passing to the new variables $x_j = B^{-1} p_j \lambda_j$, we further convert (PET) to the equivalent form

$$\min_x \left\{ f(x) \equiv - \sum_i y_i \ln(\sum_j q_{ij} x_j) : x \in \Delta_n \right\}, \quad (\text{PET}')$$

$$\left[q_{ij} = \frac{B p_{ij}}{p_j} \right]$$

which is a problem of minimizing a convex function over the standard simplex.

Another example is ℓ_1 minimization arising in sparsity-oriented signal processing, in particular, in Compressed Sensing, see section 1.3.1. The optimization problems arising here can be reduced to small series of problems of the form

$$\min_{x: \|x\|_1 \leq 1} \|Ax - b\|_p, \quad [A \in \mathbf{R}^{m \times n}]$$

where $p = \infty$ or $p = 2$. Representing $x \in \mathbf{R}^n$ as $u - v$ with nonnegative u, v , we can rewrite this problem equivalently as

$$\min_{u, v \geq 0, \sum_i u_i + \sum_i v_i = 1} \|A[u - v] - b\|_p;$$

the domain of this problem is the standard $2n$ -dimensional simplex Δ_{2n} .

Intermediate conclusion. The discussion above says that this perhaps is a good idea to look for simple convex minimization techniques which, as applied to convex programs (CP) with feasible sets of appropriate geometry, exhibit *dimension-independent* (or nearly dimension-independent) and nearly optimal information-based complexity. We are about to present a family of techniques of this type.

5.2 The Simplest: Subgradient Descent and Euclidean Bundle Level

5.2.1 Subgradient Descent

The algorithm. The “simplest of the simple” algorithm or nonsmooth convex optimization – *Subgradient Descend* (SD) discovered by N. Shor in 1967 is aimed at solving a convex program

$$\min_{x \in X} f(x), \quad (\text{CP})$$

where

- X is a convex compact set in \mathbf{R}^n , and
- f is a Lipschitz continuous on X convex function.

SD is the recurrence

$$x_{t+1} = \Pi_X(x_t - \gamma_t f'(x_t)) \quad [x_1 \in X] \quad (\text{SD})$$

where

- $\gamma_t > 0$ are stepsizes
- $\Pi_X(x) = \operatorname{argmin}_{y \in X} \|x - y\|_2^2$ is the standard *metric projector* on X , and
- $f'(x)$ is a *subgradient* of f at x :

$$f(y) \geq f(x) + (y - x)^T f'(x) \quad \forall y \in X. \quad (5.2.1)$$

Note: We always assume that $\operatorname{int} X \neq \emptyset$ and that the subgradients $f'(x)$ reported by the First Order oracle at points $x \in X$ satisfy the requirement

$$f'(x) \in \operatorname{cl} \{f'(y) : y \in \operatorname{int} X\}.$$

With this assumption, for every norm $\|\cdot\|$ on \mathbf{R}^n and for every $x \in X$ one has

$$\|f'(x)\|_* \equiv \max_{\xi: \|\xi\| \leq 1} \xi^T f'(x) \leq L_{\|\cdot\|}(f) \equiv \sup_{\substack{x \neq y, \\ x, y \in X}} \frac{|f(x) - f(y)|}{\|x - y\|}. \quad (5.2.2)$$

When, why and how SD converges? To analyse convergence of SD, we start with a simple geometric fact:

(!) Let $X \subset \mathbf{R}^n$ be a closed convex set, and let $x \in \mathbf{R}^n$. Then the vector $e = x - \Pi_X(x)$ forms an acute angle with every vector of the form $y - \Pi_X(x)$, $y \in X$:

$$(x - \Pi_X(x))^T (y - \Pi_X(x)) \leq 0 \quad \forall y \in X. \quad (5.2.3)$$

In particular,

$$y \in X \Rightarrow \|y - \Pi_X(x)\|_2^2 \leq \|y - x\|_2^2 - \|x - \Pi_X(x)\|_2^2 \quad (5.2.4)$$

Indeed, when $y \in X$ and $0 \leq t \leq 1$, one has

$$\phi(t) = \underbrace{\|\Pi_X(x) + t(y - \Pi_X(x))\|_2^2}_{y_t \in X} - x \|_2^2 \geq \|\Pi_X(x) - x\|_2^2 = \phi(0),$$

whence

$$0 \leq \phi'(0) = 2(\Pi_X(x) - x)^T(y - \Pi_X(x)).$$

Consequently,

$$\begin{aligned} \|y - x\|_2^2 &= \|y - \Pi_X(x)\|_2^2 + \|\Pi_X(x) - x\|_2^2 + 2(y - \Pi_X(x))^T(\Pi_X(x) - x) \\ &\geq \|y - \Pi_X(x)\|_2^2 + \|\Pi_X(x) - x\|_2^2. \end{aligned}$$

Corollary 5.2.1 *With x_t defined by (SD), for every $u \in X$ one has*

$$\gamma_t(x_t - u)^T f'(x_t) \leq \underbrace{\frac{1}{2}\|x_t - u\|_2^2}_{d_t} - \underbrace{\frac{1}{2}\|x_{t+1} - u\|_2^2}_{d_{t+1}} + \frac{1}{2}\gamma_t^2\|f'(x_t)\|_2^2. \quad (5.2.5)$$

Indeed, by (!) we have

$$d_{t+1} \leq \frac{1}{2}\|[x_t - u] - \gamma_t f'(x_t)\|_2^2 = d_t - \gamma_t(x_t - u)^T f'(x_t) + \frac{1}{2}\gamma_t^2\|f'(x_t)\|_2^2. \quad \square$$

Summing up inequalities (5.2.5) over $t = T_0, T_0 + 1, \dots, T$, we get

$$\begin{aligned} \sum_{t=T_0}^T \gamma_t(f(x_t) - f(u)) &\leq \underbrace{d_{T_0} - d_{T+1}}_{\leq \Theta} + \sum_{t=T_0}^T \frac{1}{2}\gamma_t^2\|f'(x_t)\|_2^2 \\ &\left[\Theta = \max_{x,y \in X} \frac{1}{2}\|x - y\|_2^2 \right] \end{aligned}$$

Setting $u = x_* \equiv \operatorname{argmin}_X f$, we arrive at the bound

$$\forall (T, T_0, T \geq T_0 \geq 1) : \epsilon_T \equiv \min_{t \leq T} f(x_t) - f_* \leq \frac{\Theta + \frac{1}{2} \sum_{t=T_0}^T \gamma_t^2\|f'(x_t)\|_2^2}{\sum_{t=T_0}^T \gamma_t} \quad (5.2.6)$$

Note that ϵ_T is the non-optimality, in terms of f , of the best (with the smallest value of f) solution x_{bst}^T found in course of the first T steps of SD.

Relation (5.2.6) allows to arrive at various convergence results.

Example 1: “Divergent Series”. Let $\gamma_t \rightarrow 0$ as $t \rightarrow \infty$, while $\sum_t \gamma_t = \infty$. Then

$$\lim_{T \rightarrow \infty} \epsilon_T = 0. \quad (5.2.7)$$

Proof. Set $T_0 = 1$ and note that

$$\frac{\sum_{t=1}^T \gamma_t^2\|f'(x_t)\|_2^2}{\sum_{t=1}^T \gamma_t} \leq L_{\|\cdot\|_2}^2(f) \frac{\sum_{t=1}^T \gamma_t^2}{\sum_{t=1}^T \gamma_t} \rightarrow 0, \quad T \rightarrow \infty.$$

Example 2: “Optimal stepsizes”: When

$$\gamma_t = \frac{\sqrt{2\Theta}}{\|f'(x_t)\|\sqrt{t}} \quad (5.2.8)$$

one has

$$\epsilon_T \equiv \min_{t \leq T} f(x_t) - f_* \leq O(1) \frac{L_{\|\cdot\|_2}(f)\sqrt{\Theta}}{\sqrt{T}}, \quad T \geq 1 \quad (5.2.9)$$

Proof. Setting $T_0 = \lfloor T/2 \rfloor$, we get

$$\begin{aligned} \epsilon_T &\leq \frac{\Theta + \Theta \sum_{t=T_0}^T \frac{1}{t}}{\sum_{t=T_0}^T \frac{\sqrt{\Theta}}{\sqrt{t}\|f'(x_t)\|_2}} \leq \frac{\Theta + \Theta \sum_{t=T_0}^T \frac{1}{t}}{\sum_{t=T_0}^T \frac{\sqrt{\Theta}}{\sqrt{t}L_{\|\cdot\|_2}(f)}} \leq L_{\|\cdot\|_2}(f)\sqrt{\Theta} \frac{1+O(1)}{O(1)\sqrt{T}} = O(1) \frac{L_{\|\cdot\|_2}(f)\sqrt{\Theta}}{\sqrt{T}} \end{aligned}$$

Good news: We have arrived at efficiency estimate which is *dimension-independent*, provided that the “ $\|\cdot\|_2$ -variation” of the objective on the feasible domain

$$\text{Var}_{\|\cdot\|_2, X}(f) = L_{\|\cdot\|_2}(f) \max_{x, y \in X} \|x - y\|_2$$

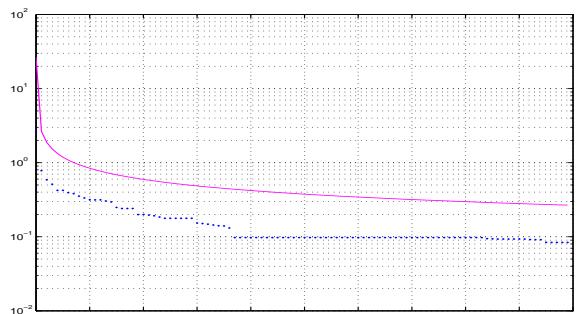
is fixed. Moreover, when X is a Euclidean ball in \mathbf{R}^n , this efficiency estimate “is as good as an efficiency estimate of a black-box-oriented method can be”, provided that the dimension is large:

$$n \geq \left(\frac{\text{Var}_{\|\cdot\|_2, X}(f)}{\epsilon} \right)^2$$

Bad news: Our “dimension-independent” efficiency estimate

- is pretty slow
- is indeed dimension-independent only for problems with “Euclidean geometry” – those with moderate $\|\cdot\|_2$ -variation. As a matter of fact, in applications problems of this type are pretty rare.

A typical “convergence pattern” is exhibited on the plot below:



SD as applied to $\min_{\|x\|_2 \leq 1} \|Ax - b\|_1$, $A : 50 \times 50$
[red: efficiency estimate; blue: actual error]

We see that after rapid progress at the initial iterates, the method “gets stuck,” reflecting the slow convergence rate.

5.2.2 Incorporating memory: Euclidean Bundle Level Algorithm

An evident drawback of SD is that all information on the objective accumulated so far is “summarized” in the current iterate, and this “summary” is very incomplete. With better usage of past information, one arrives at *bundle methods* which outperform SD significantly in practice, while preserving the most attractive theoretical property of SD – dimension-independent and optimal, in favourable circumstances, rate of convergence.

Bundle-Level algorithm. The Bundle Level algorithms works as follows.

- At the beginning of step t of BL, we have at our disposal
 - the first-order information $\{f(x_\tau), f'(x_\tau)\}_{1 \leq \tau < t}$ on f along the previous search points $x_\tau \in X$, $\tau < t$;
 - current iterate $x_t \in X$.
- At step t we
 1. compute $f(x_t), f'(x_t)$; this information, along with the past first-order information on f , provides is with the current *model of the objective*

$$f_t(x) = \max_{\tau \leq t} [f(x_\tau) + (x - x_\tau)^T f'(x_\tau)]$$

This model underestimates the objective and is exact at the points x_1, \dots, x_t ;

2. define the *best found so far value of the objective* $f^t = \min_{\tau \leq t} f(x_\tau)$
3. define the current *lower bound* f_t on f_* by solving the auxiliary problem

$$f_t = \min_{x \in X} f_t(x) \quad (\text{LP}_t)$$

Note that the current *gap* $\Delta_t = f^t - f_t$ is an upper bound on the inaccuracy of the best found so far approximate solution to the problem;

4. compute the current *level* $\ell_t = f_t + \lambda \Delta_t$ ($\lambda \in (0, 1)$ is a parameter)
5. build a new search point by solving the auxiliary problem

$$x_{t+1} = \operatorname{argmin}_x \{\|x - x_t\|_2^2 : x \in X, f_t(x) \leq \ell_t\} \quad (\text{QP}_t)$$

and loop to step $t + 1$.

Why and how BL converges? Let us start with several observations.

- The models $f_t(x) = \max_{\tau \leq t} [f(x_\tau) + (x - x_\tau)^T f'(x_\tau)]$ grow with t and underestimate f , while the best found so far values of the objective decrease with t and overestimate f_* . Thus,

$$\begin{aligned} f_1 &\leq f_2 \leq f_3 \leq \dots \leq f_* \\ f^1 &\geq f^2 \geq f^3 \leq \dots \geq f_* \\ \Delta_1 &\geq \Delta_2 \geq \dots \geq 0 \end{aligned}$$

- Let us say that a group of subsequent iterations $J = \{s, s+1, \dots, r\}$ form a *segment*, if $\Delta_r \geq (1-\lambda)\Delta_s$. We claim that

If $J = \{s, s+1, \dots, r\}$ is a segment, then

(i) All the sets $L_t = \{x \in X : f_t(x) \leq \ell_t\}$, $t \in J$, have a point in common, specifically, (any) minimizer u of $f_r(\cdot)$ over X ;

(ii) For $t \in J$, one has $\|x_t - x_{t+1}\|_2 \geq \frac{(1-\lambda)\Delta_r}{L_{\|\cdot\|_2(f)}}$.

Indeed,

(i): for $t \in J$ we have

$$\begin{aligned} f_t(u) &\leq f_r(u) = f_r = f^r - \Delta_r \leq f^t - \Delta_r \leq f^t - (1-\lambda)\Delta_s \\ &\leq f^t - (1-\lambda)\Delta_t = \ell_t. \end{aligned}$$

(ii): We have $f_t(x_t) = f(x_t) \geq f^t$, and $f_t(x_{t+1}) \leq \ell_t = f^t - (1-\lambda)\Delta_t$. Thus, when passing from x_t to x_{t+1} , t -th model decreases by at least $(1-\lambda)\Delta_t \geq (1-\lambda)\Delta_r$. It remains to note that $f_t(\cdot)$ is Lipschitz continuous w.r.t. $\|\cdot\|_2$ with constant $L_{\|\cdot\|_2}(f)$.

• **Main observation:**

(!) The cardinality of a segment $J = \{s, s+1, \dots, r\}$ of iterations can be bounded as follows:

$$\text{Card}(J) \leq \frac{\text{Var}_{\|\cdot\|_2, X}^2(f)}{(1-\lambda)^2 \Delta_r^2}.$$

Indeed, when $t \in J$, the sets $L_t = \{x \in X : f_t(x) \leq \ell_t\}$ have a point u in common, and x_{t+1} is the projection of x_t onto L_t . It follows that

$$\begin{aligned} \|x_{t+1} - u\|_2^2 &\leq \|x_t - u\|_2^2 - \|x_t - x_{t+1}\|_2^2 \quad \forall t \in J \\ \Rightarrow \sum_{t \in J} \|x_t - x_{t+1}\|_2^2 &\leq \|x_s - u\|_2^2 \leq \max_{x, y \in X} \|x - y\|_2^2 \\ \Rightarrow \text{Card}(J) &\leq \frac{\max_{x, y \in X} \|x - y\|_2^2}{\min_{t \in J} \|x_t - x_{t+1}\|_2^2} \\ \Rightarrow \text{Card}(J) &\leq \frac{L_{\|\cdot\|_2}^2(f) \max_{x, y \in X} \|x - y\|_2^2}{(1-\lambda)^2 \Delta_r^2} \quad [\text{by (ii)}] \end{aligned}$$

Corollary 5.2.2 For every ϵ , $0 < \epsilon < \Delta_1$, the number N of steps before a gap $\leq \epsilon$ is obtained (i.e., before an ϵ -solution is found) does not exceed the bound

$$N(\epsilon) = \frac{\text{Var}_{\|\cdot\|_2, X}^2(f)}{\lambda(1-\lambda)^2(2-\lambda)\epsilon^2}. \quad (5.2.10)$$

Proof. Assume that N is such that $\Delta_N > \epsilon$, and let us bound N from above. Let us split the set of iterations $I = \{1, \dots, N\}$ into segments J_1, \dots, J_m as follows:

J_1 is the maximal segment which ends with iteration N :

$$J_1 = \{t : t \leq N, (1 - \lambda)\Delta_t \leq \Delta_N\}$$

J_1 is certain group of subsequent iterations $\{s_1, s_1 + 1, \dots, N\}$. If J_1 differs from I : $s_1 > 1$, we define J_2 as the maximal segment which ends with iteration $s_1 - 1$:

$$J_2 = \{t : t \leq s_1 - 1, (1 - \lambda)\Delta_t \leq \Delta_{s_1-1}\} = \{s_2, s_2 + 1, \dots, s_1 - 1\}$$

If $J_1 \cup J_2$ differs from I : $s_2 > 1$, we define J_3 as the maximal segment which ends with iteration $s_2 - 1$:

$$J_3 = \{t : t \leq s_2 - 1, (1 - \lambda)\Delta_t \leq \Delta_{s_2-1}\} = \{s_3, s_3 + 1, \dots, s_2 - 1\}$$

and so on.

As a result, I will be partitioned “from the end to the beginning” into segments of iterations J_1, J_2, \dots, J_m . Let d_ℓ be the gap corresponding to the last iteration from J_ℓ . By maximality of segments J_ℓ , we have

$$\begin{aligned} d_1 &\geq \Delta_N > \epsilon \\ d_{\ell+1} &> (1 - \lambda)^{-1}d_\ell, \quad \ell = 1, 2, \dots, m - 1 \end{aligned}$$

whence

$$d_\ell > \epsilon(1 - \lambda)^{-(\ell-1)}.$$

We now have

$$\begin{aligned} N &= \sum_{\ell=1}^m \text{Card}(J_\ell) \leq \sum_{\ell=1}^m \frac{\text{Var}_{\|\cdot\|_2, X}^2(f)}{(1-\lambda)^2 d_\ell^2} \leq \frac{\text{Var}_{\|\cdot\|_2, X}^2(f)}{(1-\lambda)^2} \sum_{\ell=1}^m (1-\lambda)^{2(\ell-1)} \epsilon^{-2} \\ &\leq \frac{\text{Var}_{\|\cdot\|_2, X}^2(f)}{(1-\lambda)^2 \epsilon^2} \sum_{\ell=1}^{\infty} (1-\lambda)^{2(\ell-1)} = \frac{\text{Var}_{\|\cdot\|_2, X}^2(f)}{(1-\lambda)^2 [1-(1-\lambda)^2] \epsilon^2} = N(\epsilon). \end{aligned} \quad \square$$

Discussion. 1. We have seen that the Bundle-Level method shares the dimension-independent (and optimal in the “favourable” large-scale case) theoretical complexity bound

For every $\epsilon > 0$, the number of steps before an ϵ -solution to convex program $\min_{x \in X} f(x)$ is found, does not exceed $O(1) \left(\frac{\text{Var}_{\|\cdot\|_2, X}^2(f)}{\epsilon} \right)^2$.

There exists quite convincing experimental evidence that the Bundle-Level method obeys the optimal in fixed dimension “polynomial time” complexity bound

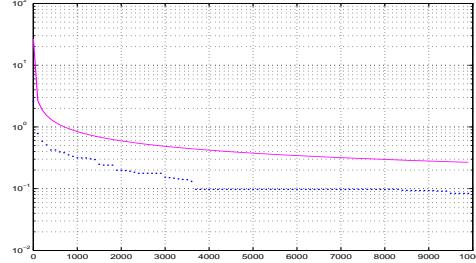
For every $\epsilon \in (0, \text{Var}_X(f) \equiv \max_X f - \min_X f)$, the number of steps before an ϵ -solution to convex program $\min_{x \in X \subset \mathbf{R}^n} f(x)$ is found does not exceed $n \ln \left(\frac{\text{Var}_X(f)}{\epsilon} \right) + 1$.

Experimental rule: When solving convex program with n variables by BL, every n steps add new accuracy digit.

Illustration: Consider the problem

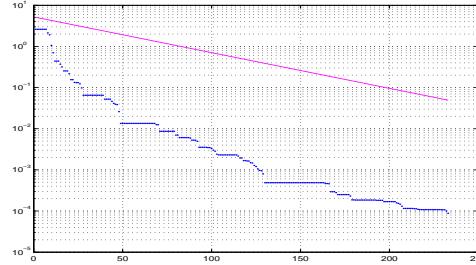
$$f_* = \min_{x: \|x\|_2 \leq 1} f(x) \equiv \|Ax - b\|_1,$$

with somehow generated 50×50 matrix A ; in the problem, $f(0) = 2.61$, $f_* = 0$. This is how SD and BL work on this problem



SD, accuracy vs. iteration count

$$\text{blue: errors; red: efficiency estimate } 3 \frac{\text{Var}_{\|\cdot\|_2, X}(f)}{\sqrt{t}} \\ \epsilon_{10000} = 0.084$$



BL, accuracy vs. iteration count

$$\text{(blue: errors; red: efficiency estimate } e^{-\frac{t}{n}} \text{Var}_X(f)) \\ \epsilon_{233} < 1.e-4$$

Bundle-Level vs. Subgradient Descent

2. In BL, the number of linear constraints in the auxiliary problems

$$f_t = \min_{x \in X} f_t(x) \quad (\text{LP}_t)$$

$$x_{t+1} = \operatorname{argmin}_x \{ \|x_t - x\|_2^2 : x \in X, f_t(x) \leq \ell_t \} \quad (\text{QP}_t)$$

is equal to the size t of the current *bundle* – the collection of affine forms $g_\tau(x) = f(x_\tau) + (x - x_\tau)^T f'(x_\tau)$ participating in model $f_t(\cdot)$. Thus, the complexity of an iteration in BL grows with the iteration number. In order to suppress this phenomenon, one needs a mechanism for *shrinking* the bundle (and thus – simplifying the models of f).

- The simplest way of shrinking the bundle is to initialize d as Δ_1 and to run plain BL until an iteration t with $\Delta_t \leq d/2$ is met. At such an iteration, we

- shrink the current bundle, keeping in it the minimum number of the forms g_τ sufficient to ensure that

$$f_t \equiv \min_{x \in X} \max_{1 \leq \tau \leq t} g_\tau(x) = \min_{x \in X} \max_{\text{selected } \tau} g_\tau(x)$$

(this number is at most n), and

- reset d as Δ_t ,

and proceed with plain BL until the gap is again reduced by factor 2, etc.

Computational experience demonstrates that the outlined approach does not slow BL down, while keeping the size of the bundle below the level of about $2n$.

What is ahead. Subgradient Descent looks very geometric, which simplifies “digesting” of the construction. However, on a close inspection this “geometricity” *is completely misleading*; the “actual” reasons for convergence have nothing to do with projections, distances, etc. What in fact is going on, is essentially less transparent and will be explained in the next section.

5.3 Mirror Descent algorithm

5.3.1 Problem and assumptions

In the sequel, we focus on the problem

$$\min_{x \in X} f(x) \quad (\text{CP})$$

where X is a closed convex set in a Euclidean space E with inner product $\langle \cdot, \cdot \rangle$ (unless otherwise is explicitly stated, we assume w.l.o.g. E to be just \mathbf{R}^n) and assume that

(A.1): *The (convex) objective f is Lipschitz continuous on X .*

To quantify this assumption, we fix once for ever a norm $\|\cdot\|$ on E and associate with f the Lipschitz constant of $f|_X$ w.r.t. the norm $\|\cdot\|$:

$$L_{\|\cdot\|}(f) = \min \{L : |f(x) - f(y)| \leq L\|x - y\| \forall x, y \in X\}.$$

Note that from Convex Analysis it follows that f at every point $x \in X$ admits a subgradient $f'(x)$ such that

$$\|f'(x)\|_* \leq L_{\|\cdot\|}(f),$$

where $\|\cdot\|_*$ is the norm conjugate to $\|\cdot\|$:

$$\|\xi\|_* = \max\{\langle \xi, x \rangle : \|x\| \leq 1\}.$$

For example, the norm conjugate to $\|\cdot\|_p$ is $\|\cdot\|_q$, where $q = \frac{p}{p-1}$.

We assume that this “small norm” subgradient $f'(x)$ is exactly the one reported by the First Order oracle as called with input $x \in X$:

$$\|f'(x)\|_* \leq L_{\|\cdot\|}(f) \quad \forall x \in X; \quad (5.3.1)$$

this is not a severe restriction, since at least in the interior of X all subgradients of f are “small” in the outlined sense.

Note that by the definition of the conjugate norm, we have

$$|\langle x, \xi \rangle| \leq \|x\| \|\xi\|_* \quad \forall x, \xi \in \mathbf{R}^n. \quad (5.3.2)$$

5.3.2 Proximal setup

The setup for the forthcoming generic algorithms for solving (CP) is given by

1. The domain X of the problem,
2. A norm $\|\cdot\|$ on the space \mathbf{R}^n embedding X ; this is the norm participating in (A.1);
3. A *distance generating function* (DGF for short) $\omega(x) : X \rightarrow \mathbf{R}$ — a convex and continuous function on X with the following properties:
 - ω admits a *continuous in $x \in X^o$* selection $\omega'(x)$ of subgradients¹; here X^o is the set of all points $x \in X$ where ω admits a subgradient;
 - $\omega(x)$ is *compatible with $\|\cdot\|$* , meaning that $\omega(x)$ is strongly convex, modulus 1, w.r.t. the norm $\|\cdot\|$, that is,

$$\forall x, y \in X^o : \langle \omega'(x) - \omega'(y), x - y \rangle \geq \|x - y\|^2,$$

or, equivalently,

$$\omega(y) \geq \omega(x) + \langle \omega'(x), y - x \rangle + \frac{1}{2} \|x - y\|^2 \quad \forall (x \in X^o, y \in X). \quad (5.3.3)$$

The outlined setup (which we refer to as *Proximal setup*) specifies several other important entities, specifically

- The ω -center of X — the point

$$x_\omega = \underset{x \in X}{\operatorname{argmin}} \omega(x).$$

This point does exist (since X is closed and convex, and ω is continuous and strongly convex on X) and is unique due to the strong convexity of ω . Besides this, $x_\omega \in X^o$ and

$$\langle \omega'(x_\omega), x - x_\omega \rangle \geq 0 \quad \forall x \in X \quad (5.3.4)$$

due to the following immediate

Fact 5.3.1 *With our ω , X^o is exactly the set of minimizers, over X , of functions of the form $\langle p, x \rangle + \omega(x)$. Besides this, the latter function attains its minimum at a point $\bar{x} \in X$ if and only if*

$$\langle \omega'(\bar{x}) + p, x - \bar{x} \rangle \geq 0 \quad \forall x \in X. \quad (5.3.5)$$

For proof, see section 5.6.1.

Note that Fact 5.3.1 can be strengthen to the following

Fact 5.3.2 *Let X , X^o , $\|\cdot\|$, $\omega(\cdot)$ be as above and let U be a closed convex subset of X which intersects the relative interior of X . Then for every p , the minimizer x_* of $h_p(x) = \langle p, x \rangle + \omega(x)$ over U exists and is unique, belongs to X^o and is fully characterized by the relations $x_* \in U \cap X^o$ and*

$$\langle p + \omega'(x_*), u - x_* \rangle \geq 0 \quad \forall u \in U. \quad (5.3.6)$$

¹From now on, “subgradient” means “subgradient on X ”

For proof, see section 5.6.1.

- The prox-term (or *local distance*, or *Bregman distance* from x to y), defined as

$$V_x(y) = \omega(y) - \langle \omega'(x), y - x \rangle - \omega(x); \quad [x \in X^o, y \in X]$$

note that $0 = V_x(x)$ and $V_x(y) \geq \frac{1}{2}\|y - x\|^2$ by (5.3.3);

- The quantity

$$\tilde{\Theta} = \sup_{y \in X} V_{x_\omega}(y) = \sup_{y \in X} [\omega(y) - \langle \omega'(x_\omega), y - x_\omega \rangle - \omega(x_\omega)] \leq \Theta := \sup_X \omega(\cdot) - \min_X \omega(\cdot),$$

where the concluding inequality is due to (5.3.4). Every one of the quantities $\tilde{\Theta}$, Θ is finite if and only if X is bounded. For a bounded X , the quantity

$$\Omega = \sqrt{2\tilde{\Theta}},$$

will be called the *ω -radius* of X . Observe that from $V_x(y) \geq \frac{1}{2}\|y - x\|^2$ it follows that $\|y - x\|^2 \leq 2V_x(y)$ whenever $x \in X^o$, $y \in X$, which combines with the definitions of $\tilde{\Theta}$ and Ω to imply that

$$\tilde{\Theta} < \infty \Rightarrow \|y - x_\omega\| \leq \Omega \quad \forall y \in X; \quad (5.3.7)$$

this is where the name of Ω comes from.

Remark 5.3.1 The importance of ω -radius of a domain X stems from the fact that, as we shall see in the man time, *the efficiency estimate* (the dependence of accuracy on iteration count) of *Mirror Descent algorithm as applied to problem (CP) on a bounded domain X depends on the underlying proximal setup solely via the value Ω of the ω -radius of X (the less is this radius, the better).*

- Prox-mapping

$$\text{Prox}_x(\xi) = \operatorname{argmin}_{y \in X} \{V_x(y) + \langle \xi, y \rangle\} = \operatorname{argmin}_{y \in X} \{\omega(y) + \langle \xi - \omega'(x), y \rangle\} : \mathbf{R}^n \rightarrow X.$$

Here the parameter of the mapping — *prox-center* x — is a point from X^o . By Fact 5.3.1, the prox-mapping takes its values in X^o .

Sometimes we shall use a modification of the prox-mapping as follows:

$$\text{Prox}_x^U(\xi) = \operatorname{argmin}_{y \in U} \{V_x(y) + \langle \xi, y \rangle\} = \operatorname{argmin}_{y \in U} \{\omega(y) + \langle \xi - \omega'(x), y \rangle\} : \mathbf{R}^n \rightarrow U$$

where U is a closed convex subset of X intersecting the relative interior of X . Here the prox-center x is again a point from X^o . By Fact 5.3.2, the latter mapping takes its values in $U \cap X^o$.

Note that the methods we are about to present require at every step computing one (or two) values of the prox-mapping, and in order for the iterations to be simple, this task should be relatively easy. In other words, from the viewpoint of implementation, we need X and ω to be “simple” and to “match each other.”

5.3.3 Standard proximal setups

We are about to list the standard proximal setups we will be working with. Justification of these setups (i.e., verifying that the DGFs ω listed below are compatible with the corresponding norms, and that the ω -radii of the domains to follow admit the bounds we are about to announce) is relegated to section 5.6

5.3.3.1 Ball setup

The Ball (called also *the Euclidean*) setup is as follows: E is a Euclidean space, $\|x\| = \|x\|_2 := \sqrt{\langle x, x \rangle}$, $\omega(x) = \frac{1}{2}\langle x, x \rangle$.

Here $X^o = X$, $V_x(y) = \frac{1}{2}\|x - y\|_2^2$, x_ω is the $\|\cdot\|_2$ -closest to the origin point of X . For a bounded X , we have $\Omega = \max_{y \in X} \|y - x_\omega\|_2$. The prox-mapping is

$$\text{Prox}_x(\xi) = \Pi_X(x - \xi),$$

where

$$\Pi_X(h) = \operatorname{argmin}_{y \in X} \|h - y\|_2$$

is what is called the *metric projector onto X* . This projector is easy to compute when X is $\|\cdot\|_p$ -ball, or the “nonnegative part” of $\|\cdot\|_p$ -ball – the set $\{x \geq a : \|x - a\|_p \leq r\}$, or the *standard simplex*

$$\Delta_n = \{x \in \mathbf{R}^n : x \geq 0, \sum_i x_i = 1\},$$

of the *full-dimensional standard simplex*

$$\Delta_n^+ = \{x \in \mathbf{R}^n : x \geq 0, \sum_i x_i \leq 1\}.$$

Finally, the norm conjugate to $\|\cdot\|_2$ is $\|\cdot\|_2$ itself.

5.3.3.2 Entropy setup

For Entropy setup, $E = \mathbf{R}^n$, $\|\cdot\| = \|\cdot\|_1$, $\omega(x)$ is the *entropy*

$$\omega(x) = \sum_{i=1}^n x_i \ln(x_i) : \Delta_n \rightarrow \mathbf{R}. \quad (5.3.8)$$

We shall use this setup *only when X is a convex and closed subset of the full-dimensional simplex Δ_n^+ with $n \geq 3$ and, in addition, X contains a positive vector*, in which case $\omega(\cdot)$ indeed is strongly convex, modulus 1 w.r.t. $\|\cdot\|_1$, on X , $X^o = \{x \in X : x > 0\}$, and

$$\Omega \leq \sqrt{2 \ln(n)}. \quad (5.3.9)$$

The ω -center can be easily pointed out when X is either the standard simplex Δ_n , or the full-dimensional standard simplex Δ_n^+ ; in both cases

$$x_\omega = [1/n; \dots; 1/n].$$

In these two cases it is also easy to compute the prox-mapping; here is an explicit formula for the case of $X = \Delta_n$ (work out the case of $X = \Delta_n^+$ by yourself):

$$X = \Delta_n \Rightarrow \text{Prox}_x(\xi) = \left[\sum_i x_i e^{-\xi_i} \right]^{-1} [x_1 e^{-\xi_1}; \dots; x_n e^{-\xi_n}].$$

The norm conjugate to $\|\cdot\|_1$ is the uniform norm $\|\cdot\|_\infty$.

5.3.3.3 ℓ_1/ℓ_2 and Simplex setups

Now let $E = \mathbf{R}^{k_1+\dots+k_n} = \mathbf{R}^{k_1} \times \dots \times \mathbf{R}^{k_n}$, so that $x \in E$ can be represented as a block vector with n blocks $x^i[x] \in \mathbf{R}^{k_i}$. The norm $\|\cdot\|$ is the so called ℓ_1/ℓ_2 , or *block- ℓ_1* , norm:

$$\|x\| = \sum_{i=1}^n \|x^i[x]\|_2.$$

Bounded case. When X is a closed convex subset of the unit ball $Z = \{x \in E : \|x\| \leq 1\}$, which we refer to as *bounded case*², a good choice of ω is

$$\omega(x) = \frac{1}{p\gamma} \sum_{j=1}^p \|x^j\|_2^p, \quad p = \begin{cases} 2, & n \leq 2 \\ 1 + \frac{1}{\ln n}, & n \geq 3 \end{cases}, \quad \gamma = \begin{cases} 1, & n = 1 \\ \frac{1}{2}, & n = 2 \\ \frac{1}{e \ln(n)}, & n > 2 \end{cases} \quad (5.3.10)$$

which results in

$$\Omega \leq O(1) \sqrt{\ln(n+1)}. \quad (5.3.11)$$

General case. It turns out that the function

$$\hat{\omega}(x = [x^1; \dots; x^n]) = \frac{n^{(p-1)(2-p)/p}}{2\gamma} \left[\sum_{j=1}^n \|x^j\|_2^p \right]^{\frac{2}{p}} : E \rightarrow \mathbf{R} \quad (5.3.12)$$

with p, γ given by (5.3.10) is a compatible with $\|\cdot\|$ -norm DGF for the entire E , so that its restriction on an arbitrary closed convex nonempty subset X of E is a DGF for X compatible with $\|\cdot\|$. When, in addition, X is bounded, the $\hat{\omega}|_X$ -radius of X can be bounded as

$$\Omega \leq O(1) \sqrt{\ln(n+1)} R, \quad (5.3.13)$$

where $R < \infty$ is such that $X \subset \{x \in E : \|x\| \leq R\}$.

In both Bounded and General cases, $X^o = X$. Computing the prox-mapping is easy when, e.g.,

- A. $X = \{x : \|x\| \leq 1\}$ and the DGF (5.3.10) is used, or
- B. $X = E$ and the DGF (5.3.12) is used.

Indeed, in the case of A, computing prox-mapping reduces to solving convex program of the form

$$\min_{y^1, \dots, y^n} \left\{ \sum_{j=1}^n [\langle y^j, b^j \rangle + \|y^j\|_2^p : \sum_j \|y^j\|_2 \leq 1] \right\}; \quad (P)$$

²note that whenever X is bounded, we can make X a subset of B_{ℓ_1/ℓ_2} by shift and scaling.

clearly, at the optimum y^j are nonpositive multiples of b^j , which immediately reduces the problem to

$$\min_{s_1, \dots, s_n} \left\{ \sum_j [s_j^p - \beta_j s_j] : s \geq 0, \sum_j s_j \leq 1 \right\}, \quad (P')$$

where $\beta_j \geq 0$ are given and $p > 1$. The latter problem is just a problem of minimizing a single separable convex function under a single separable constraint. The Lagrange dual of this problem is just a *univariate* convex program with easy to compute (just $O(n)$ a.o.) value and derivative of the objective. As a result, (P') can be solved within machine accuracy in $O(n)$ a.o., and therefore (P) can be solved within machine accuracy in $O(\dim E)$ a.o.

In the case of B, similar argument reduces computing prox-mapping to solving the problem

$$\min_{s_1, \dots, s_n} \left\{ \| [s_1; \dots; s_n] \|_p^2 - \sum_j \beta_j s_j : s \geq 0 \right\},$$

where $\beta \geq 0$. This problem admits a closed form solution (find it!), and here again computing prox-mapping takes just $O(\dim E)$ a.o.

Comments. When $n = 1$, the ℓ_1/ℓ_2 norm becomes nothing but the Euclidean norm $\|\cdot\|_2$ on $E = \mathbf{R}^{m_1}$, and the proximal setups we have just described become nothing but the Ball setup. As another extreme, consider the case when $m_1 = \dots = m_n = 1$, that is, the ℓ_1/ℓ_2 norm becomes the plain $\|\cdot\|_1$ -norm on $E = \mathbf{R}^n$. Now let X be a closed convex subset of the unit $\|\cdot\|_1$ -ball of \mathbf{R}^n ; restricting the DGF (5.3.11) for the latter set onto X , we get what we shall call *Simplex setup*. Note that when X is a part of the full dimensional simplex Δ_n^+ , Simplex setup can be viewed as alternative to Entropy setup. Note that in the case in question both Entropy and Simplex setups yield basically identical bounds on the ω -radius of X (and thus result in algorithms with basically identical efficiency estimates, see Remark 5.3.1).

Note that in the case of $X \subset \Delta_n^+$ Simplex setup, in contrast to Entropy one, yields a DGF which is continuously differentiable on the entire X , moreover, such that

$$\sup_{\|u\|_1 \leq 1, \|v\|_1 \leq 1} V_u(v) \leq O(1) \ln(n+1) \quad (5.3.14)$$

(in fact, this bound works whenever X is contained in the unit $\|\cdot\|_1$ -ball), which is stronger than the entropy analogy of this bound, that is,

$$\sup_{v \in \Delta_n^+} V_u(v) \leq O(1) \ln(n+1)$$

where u is the barycenter of Δ_n . Note that in the entropy case, $V_u(v)$ can be arbitrarily large, provided that some entries in u are close to zero. Property (5.3.14) is quite valuable in some situations (see, e.g., the issues related to updating prox-centers on page 376). The only – and completely unimportant for all practical purposes – shortcoming of Simplex setup as compared to Entropy one is that when $X = \Delta_n$ or $X = \Delta_n^+$, prox-mapping for Entropy setup admits a closed form representation and can be computed in $O(\dim X)$ a.o., while with Simplex setup, computing prox-mapping requires solving a univariate convex program, and the same $O(\dim X)$ a.o. allow to solve the problem “only” within machine accuracy.

5.3.3.4 Nuclear norm and Spectahedron setups

The concluding pair of “standard” proximal setups we are about to consider deals with the case when E is the space of matrices. It makes sense to speak about block-diagonal matrices of a given block-diagonal structure. Specifically,

- Given a collection μ, ν of n pairs of positive integers μ_i, ν_i , let $\mathbf{M}^{\mu\nu}$ be the space of all real matrices $x = \text{Diag}\{x^1, \dots, x^n\}$ with n diagonal blocks x^i of sizes $\mu_i \times \nu_i$, $i = 1, \dots, n$. We equip this space with natural linear operations and Frobenius inner product

$$\langle x, y \rangle = \text{Tr}(xy^T).$$

The *nuclear norm* $\|\cdot\|_{\text{nuc}}$ on $\mathbf{M}^{\mu\nu}$ is defined as

$$\|x = \text{Diag}\{x^1, \dots, x^n\}\|_{\text{nuc}} = \|\sigma(x)\|_1 = \sum_{i=1}^n \|\sigma(x^i)\|_1,$$

where $\sigma(y)$ is the collection of singular values of a matrix y .

Note: We always assume that $\mu_i \leq \nu_i$ for all $i \leq n$ ³. With this agreement, we can think of $\sigma(x)$, $x \in \mathbf{M}^{\mu\nu}$, as of $m = \sum_{i=1}^n$ -dimensional vector with entries arranged in the non-ascending order.

- Given a collection ν of n positive integers ν_i , $i = 1, \dots, n$, let \mathbf{S}^ν be the space of symmetric block-diagonal matrices with n diagonal blocks of sizes $\nu_i \times \nu_i$, $i = 1, \dots, n$. $\mathbf{S}^\nu = \mathbf{S}^{\nu_1} \times \dots \times \mathbf{S}^{\nu_n}$ is just a subspace in $\mathbf{M}^{\nu\nu}$; the restriction of the nuclear norm on $\mathbf{M}^{\mu\nu}$ is sometimes called the *trace norm* on \mathbf{S}^ν ; it is nothing but the $\|\cdot\|_1$ -norm of the vector of eigenvalues of a matrix $x \in \mathbf{S}^\nu$.

Nuclear norm setup. Here $E = \mathbf{M}^{\mu\nu}$, $\mu_i \leq \nu_i$, $1 \leq i \leq n$, and $\|\cdot\|$ is the nuclear norm $\|\cdot\|_{\text{nuc}}$ on E . We set

$$m = \sum_{i=1}^n \mu_i,$$

so that m is thus the smaller size of a matrix from E .

Note that the norm $\|\cdot\|_*$ is the usual spectral norm (the largest singular value) of a matrix.

Bounded case. When X is a closed convex subset of the unit ball $Z = \{x \in E : \|x\| \leq 1\}$, which we refer to as *bounded case*, a good choice of ω is

$$\omega(x) = \frac{4\sqrt{e}\ln(2m)}{2^q(1+q)} \sum_{i=1}^m \sigma_i^{1+q}(x) : X \rightarrow \mathbf{R}, \quad q = \frac{1}{2\ln(2m)}, \quad (5.3.15)$$

which results in

$$\Omega \leq 2\sqrt{2\sqrt{e}\ln(2m)} \leq 4\sqrt{\ln(2m)}. \quad (5.3.16)$$

General case. It turns out that the function

$$\widehat{\omega}(x) = 2e\ln(2m) \left[\sum_{j=1}^m \sigma_j^{1+q}(x) \right]^{\frac{2}{1+q}} : E \rightarrow \mathbf{R}, \quad q = \frac{1}{2\ln(2m)} \quad (5.3.17)$$

³When this is not the case, we can replace “badly sized” diagonal blocks with their transposes, with no effect on the results of linear operations, inner products, and nuclear norm - the only entities we are interested in.

is a DGF for $Z = E$ compatible with $\|\cdot\|_{\text{nuc}}$, so that the restriction of $\widehat{\omega}$ on a nonempty closed and convex subset X of E is a DGF for X compatible with $\|\cdot\|_{\text{nuc}}$. When X is bounded, we have

$$\Omega \leq O(1)\sqrt{\ln(2m)}R, \quad (5.3.18)$$

where $R < \infty$ is such that $X \subset \{x \in E : \|x\|_{\text{nuc}} \leq R\}$.

The DGFs we have just defined are continuously differentiable, so that for the outlined setups, $X^o = X$. It is clear how to compute the associated prox mappings in the cases when

A. $X = \{x : \|x\| \leq 1\}$ and the DGF (5.3.15) is used, or

B. $X = E$ and the DGF (5.3.17) is used.

Indeed, let us start with the case of A, where computing the prox-mapping reduces to solving the convex program

$$\min_{y \in \mathbf{M}^{\mu\nu}} \left\{ \text{Tr}(y\eta^T) + \sum_{i=1}^m \sigma_i^p(y) : \sum_i \sigma_i(y) \leq 1 \right\} \quad (Q)$$

Let $\eta = uhv^T$ be the singular value decomposition of η , so that $u \in \mathbf{M}^{\mu\mu}$, $v \in \mathbf{M}^{\nu\nu}$ are block-diagonal orthogonal matrices, and $h \in \mathbf{M}^{\mu\nu}$ is block-diagonal matrix with diagonal blocks $h^j = [\text{Diag}\{g^j\}, 0_{\mu_j, \text{times}[\nu_j - \mu_j]}]$, $g^j \in \mathbf{R}_+^{\mu_j}$. Passing in (Q) from variable y to the variable $z = u^Tyv$, the problem becomes

$$\min_{z^j \in \mathbf{M}^{\mu_j \nu_j}, 1 \leq j \leq n} \left\{ \text{Tr}(z^j[h^j]^T) + \sum_{j=1}^n \sum_{k=1}^{\mu_j} \sigma_k^p(z^j) : \sum_{j=1}^n \sum_{k=1}^{\mu_j} \sigma_k(z^j) \leq 1 \right\}.$$

This (clearly solvable) problem has a rich group of symmetries; specifically, due to the diagonal nature of h^j , replacing a block z^j in a feasible solution with $\text{Diag}\{\epsilon\}z^j\text{Diag}\{\epsilon\}$, where ϵ is a vector of ± 1 's of dimension μ_j , and ϵ' is obtained from ϵ by adding $\nu_j - \mu_j$ of ± 1 entries to the right of ϵ . Since the problem is convex, it has an optimal solution which is preserved by the above symmetries, that is, such that all z^j are diagonal matrices. Denoting by ζ^j the diagonals of these diagonal matrices and setting $g = [g^1; \dots; g^n] \in \mathbf{R}^m$, the problem becomes

$$\min_{\zeta = [\zeta^1; \dots; \zeta^n] \in \mathbf{R}^n} \left\{ \zeta^T g + \sum_{k=1}^m |\zeta_k|^p : \sum_k |\zeta_k| \leq 1 \right\},$$

which, basically, is the problem we have met when computing prox-mapping in the case A of ℓ_1/ℓ_2 setup; as we remember, this problem can be solved within machine precision in $O(m)$ a.o.

Completely similar reasoning shows that in the case of B, computing prox-mapping reduces, at the price of a single singular value decomposition of a matrix from $\mathbf{M}^{\mu\nu}$, to solving the same problem as in the case B of ℓ_1/ℓ_2 setup. The bottom line is that *in the cases of A, B, the computational effort of computing prox-mapping is dominated by the necessity to carry out singular value decomposition of a matrix from $\mathbf{M}^{\mu\nu}$* .

Remark 5.3.2 A careful reader should have recognized at this point the deep similarity between the ℓ_1/ℓ_2 and the nuclear norm setups. This similarity has a very simple explanation: *the ℓ_1/ℓ_2 situation is a very special case of the nuclear norm one*. Indeed, a block vector $y = [y^1; \dots; y^n]$, $y^j \in \mathbf{R}^{k_j}$, can be thought of as the block-diagonal matrix y^+ with n $1 \times k_j$ diagonal blocks $[y^j]^T$, $j = 1, \dots, n$. with this identification of block-vectors y and block-diagonal matrices, the singular values of y^+ are exactly the norms $\|y^j\|_2$, $j = 1, \dots, m \equiv n$, so that the nuclear norm of y^+ is exactly the block- ℓ_1 norm of y . Up to minor differences in the coefficients in the formulas for DGFs, our proximal setups respect this reduction of the ℓ_1/ℓ_2 situation to the nuclear norm one.

Spectahedron setup. Consider a space \mathbf{S}^ν of block-diagonal symmetric matrices of block-diagonal structure ν , and let Δ_ν and Δ_ν^+ be, respectively, the standard and the full-dimensional spectahedrons in \mathbf{S}^ν :

$$\Delta_\nu = \{x \in \mathbf{S}^\nu : x \succeq 0, \text{Tr}(x) = 1\}, \quad \Delta_\nu^+ = \{x \in \mathbf{S}^\nu : x \succeq 0, \text{Tr}(x) \leq 1\}.$$

Let X be a nonempty closed convex subset of Δ_ν^+ . Restricting the DGF (5.3.15) associated with $\mathbf{M}^{\nu\nu}$ onto \mathbf{S}^ν (the latter space is a linear subspace of $\mathbf{M}^{\nu\nu}$) and further onto X , we get a continuously differentiable DGF for X , and the ω -radius of X does not exceed $O(1)\sqrt{\ln(m+1)}$, where, $m = \nu_1 + \dots + \nu_n$. When $X = \Delta_\nu$ or $X = \Delta_\nu^+$, the computational effort for computing prox-mapping is dominated by the necessity to carry out eigenvalue decomposition of a matrix from \mathbf{S}^ν .

5.3.4 Mirror Descent algorithm

5.3.4.1 Basic Fact

In the sequel, we shall use the following, crucial in our context,

Fact 5.3.3 *Given proximal setup $\|\cdot\|$, X , $\omega(\cdot)$ along with a closed convex subset U of X intersecting the relative interior of X , let $x \in X^o$, $\xi \in \mathbf{R}^n$ and $x_+ = \text{Prox}_x^U(\xi)$. Then*

$$\forall u \in U : \langle \xi, x_+ - u \rangle \leq V_x(u) - V_{x_+}(u) - V_x(x_+). \quad (5.3.19)$$

Proof. By Fact 5.3.2, we have

$$\begin{aligned} x_+ &= \operatorname{argmin}_{y \in U} \{\omega(y) + \langle \xi - \omega'(x), y \rangle\} \\ \Rightarrow \forall u \in U : \langle \omega'(x_+) - \omega'(x) + \xi, u - x_+ \rangle &\geq 0 \\ \Rightarrow \forall u \in U : \langle \xi, x_+ - u \rangle &\leq \langle \omega'(x_+) - \omega'(x), u - x_+ \rangle \\ &= [\omega(u) - \omega(x) - \langle \omega'(x), u - x \rangle] - [\omega(u) - \omega(x_+) - \langle \omega'(x_+), u - x_+ \rangle] \\ &\quad - [\omega(x_+) - \omega(x) - \langle \omega'(x), x_+ - x \rangle] \\ &= V_x(u) - V_{x_+}(u) - V_x(x_+). \quad \square \end{aligned}$$

A byproduct of the above computation is a wonderful identity:

Magic Identity: Whenever $x_+ \in X^o$, $x \in X^o$ and $u \in X$, we have

$$\langle \omega'(x_+) - \omega'(x), u - x_+ \rangle = V_x(u) - V_{x_+}(u) - V_x(x_+). \quad (5.3.20)$$

5.3.4.2 Standing Assumption

From now on, if otherwise is not explicitly stated, we assume that the domain X of problem (CP) in question is bounded.

5.3.4.3 MD: Description

The simplest version of the construction we are developing is the *Mirror Descent* (MD) algorithm for solving problem

$$\min_{x \in X} f(x). \quad (\text{CP})$$

MD is given by the recurrence

$$x_1 = x_\omega; \quad x_{t+1} = \text{Prox}_{x_t}(\gamma_t f'(x_t)), \quad t = 1, 2, \dots \quad (5.3.21)$$

where $\gamma_t > 0$ are positive stepsizes. Note that this procedure is well defined since, as we have seen, $x_\omega \in X^o$ and the values of the prox-mapping also belong to X^o , so that $x_t \in X^o$ for all t and thus $\text{Prox}_{x_t}(\cdot)$ makes sense.

The points x_t in (5.3.21) are what is called *search points* – the points where we collect the first order information on f . There is no reason to insist that these search points should be also the approximate solutions x^t generated in course of $t = 1, 2, \dots$ steps. In MD as applied to (CP), there are two good ways to define the approximate solutions x^t , specifically:

- the best found so far approximate solution x_{bst}^t – the point of the collection x_1, \dots, x_t with the smallest value of the objective f , and
- the aggregated solution

$$x^t = \left[\sum_{\tau=1}^t \gamma_\tau \right]^{-1} \sum_{\tau=1}^t \gamma_\tau x_\tau. \quad (5.3.22)$$

Note that $x^t \in X$ (as a convex combination of the points x_τ which by construction belong to X).

Remark 5.3.3 Note that MD with Ball setup is nothing but SD (check it!).

5.3.4.4 MD: Complexity analysis

Convergence properties of MD are summarized in the following simple

Theorem 5.3.1 For every $\tau \geq 1$ and every $u \in X$, one has

$$\gamma_\tau \langle f'(x_\tau), x_\tau - u \rangle \leq V_{x_\tau}(u) - V_{x_{\tau+1}}(u) + \frac{1}{2} \|\gamma_\tau f'(x_\tau)\|_*^2. \quad (5.3.23)$$

As a result, for every $t \geq 1$ one has

$$f(x^t) - \min_X f \leq \frac{\frac{\Omega^2}{2} + \frac{1}{2} \sum_{\tau=1}^t \gamma_\tau^2 \|f'(x_\tau)\|_*^2}{\sum_{\tau=1}^t \gamma_\tau}, \quad (5.3.24)$$

and similarly for x_{bst}^t in the role of x^t .

In particular, given a number T of iterations we intend to perform and setting

$$\gamma_t = \frac{\Omega}{\|f'(x_t)\|_* \sqrt{T}}, \quad 1 \leq t \leq T, \quad (5.3.25)$$

we ensure that

$$f(x^T) - \min_X f \leq \Omega \frac{\sqrt{T}}{\sum_{t=1}^T \frac{1}{\|f'(x_t)\|_*}} \leq \frac{\Omega \max_{t \leq T} \|f'(x_t)\|_*}{\sqrt{T}} \leq \frac{\Omega L_{\|\cdot\|}(f)}{\sqrt{T}}, \quad (5.3.26)$$

($L_{\|\cdot\|}(f)$ is the Lipschitz constant of f w.r.t. $\|\cdot\|$), and similarly for x_{bst}^T .

Proof. 1⁰. Let us set $\xi_\tau = \gamma_\tau f'(x_\tau)$, so that

$$x_{\tau+1} = \text{Prox}_{x_\tau}(\xi_\tau) = \operatorname{argmin}_{y \in X} [\langle \xi_\tau - \omega'(x_\tau), y \rangle + \omega(y)].$$

We have

$$\begin{aligned}
& \langle \xi_\tau - \omega'(x_\tau) + \omega'(x_{\tau+1}), u - x_{\tau+1} \rangle \geq 0 \text{ [by Fact 5.3.1]} \\
\Rightarrow & \langle \xi_\tau, x_\tau - u \rangle \leq \langle \xi_\tau, x_\tau - x_{\tau+1} \rangle + \langle \omega'(x_{\tau+1}) - \omega'(x_\tau), u - x_{\tau+1} \rangle \\
= & \langle \xi_\tau, x_\tau - x_{\tau+1} \rangle + V_{x_\tau}(u) - V_{x_{\tau+1}}(u) - V_{x_\tau}(x_{\tau+1}) \text{ [Magic Identity (5.3.20)]} \\
= & \langle \xi_\tau, x_\tau - x_{\tau+1} \rangle + V_{x_\tau}(u) - V_{x_{\tau+1}}(u) - \underbrace{[\omega(x_{\tau+1}) - \omega(x_\tau) - \langle \omega'(x_\tau), x_{\tau+1} - x_\tau \rangle]}_{\geq \frac{1}{2} \|x_{\tau+1} - x_\tau\|^2} \\
= & V_{x_\tau}(u) - V_{x_{\tau+1}}(u) + \left[\langle \xi_\tau, x_\tau - x_{\tau+1} \rangle - \frac{1}{2} \|x_{\tau+1} - x_\tau\|^2 \right] \\
\leq & V_{x_\tau}(u) - V_{x_{\tau+1}}(u) + \left[\|\xi_\tau\|_* \|x_\tau - x_{\tau+1}\| - \frac{1}{2} \|x_{\tau+1} - x_\tau\|^2 \right] \text{ [see (5.3.2)]} \\
\leq & V_{x_\tau}(u) - V_{x_{\tau+1}}(u) + \frac{1}{2} \|\xi_\tau\|_*^2 \text{ [since } ab - \frac{1}{2} b^2 \leq \frac{1}{2} a^2]
\end{aligned}$$

Recalling that $\xi_\tau = \gamma_\tau f'(x_\tau)$, we arrive at (5.3.23).

2⁰. Denoting by x_* a minimizer of f on X , substituting $u = x_*$ in (5.3.23) and taking into account that f is convex, we get

$$\gamma_\tau [f(x_\tau) - f(x_*)] \leq \gamma_\tau \langle f'(x_\tau), x_\tau - x_* \rangle \leq V_{x_\tau}(x_*) - V_{x_{\tau+1}}(x_*) + \frac{1}{2} \gamma_\tau^2 \|f'(x_\tau)\|_*^2.$$

Summing up these inequalities over $\tau = 1, \dots, t$, dividing both sides by $\Gamma_t \equiv \sum_{\tau=1}^t \gamma_\tau$ and setting $\nu_\tau = \gamma_\tau / \Gamma_t$, we get

$$\left[\sum_{\tau=1}^t \nu_\tau f(x_\tau) \right] - f(x_*) \leq \frac{V_{x_1}(x_*) - V_{x_{t+1}}(x_*) + \frac{1}{2} \sum_{\tau=1}^t \gamma_\tau^2 \|f'(x_\tau)\|_*^2}{\Gamma_t}. \quad (*)$$

Since ν_τ are positive and sum up to 1, and f is convex, the left hand side in $(*)$ is $\geq f(x^t) - f(x_*)$ (recall that $x^t = \sum_{\tau=1}^t \nu_\tau x_\tau$); it clearly is also $\geq f(x_{\text{bst}}^t) - f(x_*)$. Since $V_{x_1}(x) \leq \frac{\Omega^2}{2}$ due to the definition of Ω and to $x_1 = x_\omega$, while $V_{x_{t+1}}(\cdot) \geq 0$, the right hand side in $(*)$ is \leq the right hand side in (5.3.24), and we arrive at (5.3.24).

3⁰. Relation (5.3.26) is readily given by (5.3.24), (5.3.25) and the fact that $\|f'(x_t)\|_* \leq L_{\|\cdot\|}(f)$, see (5.3.1). \square

5.3.4.5 Refinement

Under favorable circumstances we can slightly modify the MD algorithm and refine its complexity analysis. Specifically, let us introduce the following assumption on the MD setup:

(!) In the situation and notation of sections 5.3.1, 5.3.2, for properly selected Ω_* , $0 < \Omega_* < \infty$, one has

$$\sup_{u \in X^o, v \in X} V_u(v) \leq \frac{\Omega_*^2}{2}. \quad (5.3.27)$$

Note that when X is bounded, the Ball setup, and the Bounded and General cases of ℓ_1/ℓ_2 and the Nuclear norm setups meet the above assumption. Moreover, it is easily seen that in these cases the quantities Ω_* satisfy the same, within absolute constant factor, upper bounds as the respective quantities Ω , that is, bounds (5.3.11) (ℓ_1/ℓ_2 setup, Bounded case), (5.3.13) (ℓ_1/ℓ_2 setup, General case), (5.3.16) (Nuclear norm setup, Bounded case), and (5.3.18) (Nuclear norm setup, General case). In the case of Ball setup, we clearly have $\Omega_* \leq 2\Omega$.

The outlined refinement of the MD complexity analysis is given by the following version of Theorem 5.3.1:

Theorem 5.3.2 Let (5.3.27) takes place, and let $\lambda_1, \lambda_2, \dots$ be positive reals such that

$$\lambda_1/\gamma_1 \leq \lambda_2/\gamma_2 \leq \lambda_3/\gamma_3 \leq \dots, \quad (5.3.28)$$

and let

$$x^t = \frac{\sum_{\tau=1}^t \lambda_\tau x_\tau}{\sum_{\tau=1}^t \lambda_\tau}.$$

Then $x^t \in X$ and

$$f(x^t) - \min_X f \leq \frac{\sum_{\tau=1}^t \lambda_\tau [f(x_\tau) - \min_X f]}{\sum_{\tau=1}^t \lambda_\tau} \leq \frac{\frac{\lambda_t}{\gamma_t} \Omega_*^2 + \sum_{\tau=1}^t \lambda_\tau \gamma_\tau \|f'(x_\tau)\|_*^2}{2 \sum_{\tau=1}^t \lambda_\tau}, \quad (5.3.29)$$

and similarly for x_{bst}^t in the role of x^t .

Note that (5.3.24) is nothing but (5.3.29) as applied with $\lambda_\tau = \gamma_\tau$, $1 \leq \tau \leq t$, and with Ω in the role of Ω_* .

Proof of Theorem 5.3.2. We still have at our disposal (5.3.23); multiplying both sides of this relation by λ_τ/γ_τ and summing up the resulting inequalities over $\tau = 1, \dots, t$, we get

$$\begin{aligned} & \forall (u \in X) : \\ & \sum_{\tau=1}^t \lambda_\tau \langle f'(x_\tau), x_\tau - u \rangle \leq \sum_{\tau=1}^t \frac{\lambda_\tau}{\gamma_\tau} [V_{x_\tau}(u) - V_{x_{\tau+1}}(u)] + \frac{1}{2} \sum_{\tau=1}^t \lambda_\tau \gamma_\tau \|f'(x_\tau)\|_*^2 \\ & = \underbrace{\frac{\lambda_1}{\gamma_1} V_{x_1}(u)}_{\geq 0} + \underbrace{\left[\frac{\lambda_2}{\gamma_2} - \frac{\lambda_1}{\gamma_1} \right] V_{x_2}(u)}_{\geq 0} + \underbrace{\left[\frac{\lambda_3}{\gamma_3} - \frac{\lambda_2}{\gamma_2} \right] V_{x_3}(u)}_{\geq 0} + \dots + \underbrace{\left[\frac{\lambda_t}{\gamma_t} - \frac{\lambda_{t-1}}{\gamma_{t-1}} \right] V_{x_t}(u)}_{\geq 0} \\ & - \frac{\lambda_t}{\gamma_t} V_{x_{t+1}}(u) + \frac{1}{2} \sum_{\tau=1}^t \lambda_\tau \gamma_\tau \|f'(x_\tau)\|_*^2 \leq \frac{\lambda_t}{\gamma_t} \frac{\Omega_*^2}{2} + \frac{1}{2} \sum_{\tau=1}^t \lambda_\tau \gamma_\tau \|f'(x_\tau)\|_*^2, \end{aligned}$$

where the concluding inequality is due to $0 \leq V_{x_\tau}(u) \leq \frac{\Omega_*^2}{2}$ for all τ . As a result,

$$\max_{u \in X} \Lambda_t^{-1} \sum_{\tau=1}^t \lambda_\tau \langle f'(x_\tau), x_\tau - u \rangle \leq \frac{\frac{\lambda_t}{\gamma_t} \Omega_*^2 + \sum_{\tau=1}^t \lambda_\tau \gamma_\tau}{2 \Lambda_t} \quad (5.3.30)$$

$$\left[\Lambda_t = \sum_{\tau=1}^t \lambda_\tau \right].$$

On the other hand, by convexity of f we have

$$f(x^t) - f(x_*) \leq \Lambda_t^{-1} \sum_{\tau=1}^t \lambda_\tau [f(x_\tau) - f(x_*)] \leq \Lambda_t^{-1} \sum_{\tau=1}^t \lambda_\tau \langle f'(x_\tau), x_\tau - x_* \rangle, \quad (5.3.31)$$

which combines with (5.3.30) to yield (5.3.29). Since by evident reasons

$$f(x_{\text{bst}}^t) \leq \Lambda_t^{-1} \sum_{\tau=1}^t \lambda_\tau f(x_\tau),$$

the first inequality in (5.3.31) is preserved when replacing x^t with x_{bst}^t , and the resulting inequality combines with (5.3.30) to imply the validity of (5.3.29) for x_{bst}^t in the role of x_t . \square

Discussion. As we have already mentioned, in the case of (!) Theorem 5.3.1 is “essentially covered” by Theorem 5.3.2. The latter, however, sometimes yields more information. For example, the “In particular” part of Theorem 5.3.1 states that for every given “time horizon” T , specifying γ_t , $t \leq T$, according to (5.3.25), we can upper-bound the non-optimalities $f(x_{\text{bst}}^T) - \min_X f$ and $f(x^T) - \min_X f$ of the approximate solutions generated in course of T steps by $\frac{\Omega_* L_{\|\cdot\|}(f)}{\sqrt{T}}$, (see (5.3.26)), which is the best result we can extract from Theorem; note, however, that to get this “best result,” we should tune the stepsizes to the total number of steps T we intend to perform. What to do if we do not want to fix the total number T of steps in advance? A natural way is to pass from the stepsize policy (5.3.25) to its “rolling horizon” analogy, specifically, to use

$$\gamma_t = \frac{\Omega_*}{\|f'(x_t)\|_* \sqrt{t}}, \quad t = 1, 2, \dots \quad (5.3.32)$$

With these stepsizes, the efficiency estimate (5.3.26) yields

$$\begin{aligned} \max[f(x^T), f(x_{\text{bst}}^T)] - \min_X f &\leq O(1) \frac{\Omega_* L_{\|\cdot\|}(f) \ln(T+1)}{\sqrt{T}}, \quad T = 1, 2, \dots \\ x^T &= \left[\sum_{t=1}^T \frac{\gamma_t x_t}{\sum_{t=1}^T \gamma_t} \right] \end{aligned} \quad (5.3.33)$$

(recall that $O(1)$ ’s are absolute constants); this estimate is worse than (5.3.26), although just by logarithmic in T factor⁴. On the other hand, utilizing the stepsize policy (5.3.32) and setting

$$\lambda_t = \frac{t^p}{\|f'(x_t)\|_*}, \quad t = 1, 2, \dots \quad (5.3.34)$$

where $p > -1/2$, we ensure (5.3.28) and thus can apply (5.3.29), arriving at

$$\begin{aligned} \max[f(x^t), f(x_{\text{bst}}^t)] - \min_X f &\leq C_p \frac{\Omega_* L_{\|\cdot\|}(f)}{\sqrt{t}}, \quad t = 1, 2, \dots \\ x^t &= \left[\sum_{\tau=1}^t \frac{\lambda_\tau x_\tau}{\sum_{\tau=1}^t \lambda_\tau} \right] \end{aligned} \quad (5.3.35)$$

where C_p depends solely on $p > -1/2$ and is continuous in this range of values of p . Note that the efficiency estimate (5.3.35) is, essentially, the same as (5.3.26) whatever be time horizon t . Note also that the weights with which we aggregate search points x_τ to get approximate solutions in (5.3.33) and in (5.3.35) are different: in (5.3.33), other things (specifically, $\|f'(x_\tau)\|_*$) being equal, the “newer” is a search point, the less is the weight with which the point participates in the “aggregated” approximate solution. In contrast, when $p = 0$, the aggregation (5.3.35) is “time invariant,”, and when $p > 0$, it pays the more attention to a search point the newer is the point, this feature being amplified as $p > 0$ grows. Intuitively, “more attention to the new search points” seems to be desirable, provided the search trajectory x_τ itself, not just the sequences of averaged solutions x^τ or the best found so far solutions x_{bst}^τ , converges, as $\tau \rightarrow \infty$, to the optimal set of the problem $\min_{x \in X} f(x)$; on a closest inspection, this convergence indeed takes place, provided (!) takes place and the stepsizes $\gamma_t > 0$ satisfy $\gamma_t \rightarrow 0$, $t \rightarrow \infty$, and $\sum_{t=1}^{\infty} \gamma_t = \infty$.

5.3.4.6 MD: Optimality

Now let us look what our complexity analysis says in the case of the standard setups.

⁴Another shortcoming of (5.3.33) as compared to (5.3.26) is $\Omega_* \geq \Omega$ in the role of Ω ; this is not a big deal, since for typical MD setups Ω and Ω_* are within absolute constant factor of each other.

Ball setup and optimization over the ball. As we remember, in the case of the ball setup one has

$$\Omega = \max_{y \in X} \|y - x_\omega\|_2 \leq D_{\|\cdot\|_2}(X),$$

where $D_{\|\cdot\|_2}(X) = \max_{x,y \in X} \|x - y\|_2$ if the $\|\cdot\|_2$ -diameter of X . Consequently, (5.3.26) becomes

$$f(x^t) - \min_X f \leq \frac{D_{\|\cdot\|_2}(X)L_{\|\cdot\|_2}(f)}{\sqrt{T}}, \quad (5.3.36)$$

meaning that the number $N(\epsilon)$ of MD steps needed to solve (CP) within accuracy ϵ can be bounded as

$$N(\epsilon) \leq \frac{D_{\|\cdot\|_2}^2(X)L_{\|\cdot\|_2}^2(f)}{\epsilon^2} + 1. \quad (5.3.37)$$

On the other hand, let $L > 0$, and let $\mathcal{P}_{\|\cdot\|_2,L}(X)$ be the family of all convex problems (CP) with convex Lipschitz continuous, with constant L w.r.t. $\|\cdot\|_2$, objectives. It is known that *if X is an n -dimensional Euclidean ball and $n \geq \frac{D_{\|\cdot\|_2}^2(X)L^2}{\epsilon^2}$, then the information-based complexity of the family $\mathcal{P}_{\|\cdot\|_2,L}(X)$ is at least $O(1)\frac{D_{\|\cdot\|_2}^2(X)L^2}{\epsilon^2}$* (cf. (5.1.4)). Comparing this result with (5.3.37), we conclude that

If X is an n -dimensional Euclidean ball, then the complexity of the family $\mathcal{P}_{\|\cdot\|_2,L}(X)$ w.r.t. the MD algorithm with the ball setup in the “large-scale case” $n \geq \frac{D_{\|\cdot\|_2}^2(X)L^2}{\epsilon^2}$ coincides (within an absolute constant factor) with the information-based complexity of the family.

Entropy/Simplex setup and minimization over the simplex. In the case of Entropy setup we have $\Omega \leq \sqrt{2 \ln(n)}$, so that (5.3.26) becomes

$$f(x^T) - \min_X f \leq \frac{\sqrt{2 \ln(n)}L_{\|\cdot\|_1}(f)}{\sqrt{T}}, \quad (5.3.38)$$

meaning that in order to solve (CP) within accuracy ϵ it suffices to carry out

$$N(\epsilon) \leq \frac{2 \ln(n)L_{\|\cdot\|_1}^2(f)}{\epsilon^2} + 1 \quad (5.3.39)$$

MD steps.

On the other hand, let $L > 0$, and let $\mathcal{P}_{\|\cdot\|_1,L}(X)$ be the family of all convex problems (CP) with convex Lipschitz continuous, with constant L w.r.t. $\|\cdot\|_1$, objectives. It is known that *if X is the standard n -dimensional simplex Δ_n (or the standard full-dimensional simplex Δ_n^+) and $n \geq \frac{L^2}{\epsilon^2}$, then the information-based complexity of the family $\mathcal{P}_{\|\cdot\|_1,L}(X)$ is at least $O(1)\frac{L^2}{\epsilon^2}$* (cf. (5.1.5)). Comparing this result with (5.3.37), we conclude that

If X is the n -dimensional simplex Δ_n (or the full-dimensional simplex Δ_n^+), then the complexity of the family $\mathcal{P}_{\|\cdot\|_1,L}(X)$ w.r.t. the MD algorithm with the simplex setup in the “large-scale case” $n \geq \frac{L^2}{\epsilon^2}$ coincides, within a factor of order of $\ln n$, with the information-based complexity of the family.

Note that the just presented discussion can be word by word repeated when entropy setup is replaced with Simplex setup.

Spectahedron setup and large-scale semidefinite optimization. All the conclusions we have made when speaking about the case of Entropy/Simplex setup and $X = \Delta_n$ (or $X = \Delta_n^+$) remain valid in the case of the spectahedron setup and X defined as the set of all *block-diagonal* matrices of a given block-diagonal structure contained in $\mathcal{D}_n = \{x \in \mathbf{S}^n : x \succeq 0, \text{Tr}(x) \leq 1\}$ (or contained in \mathcal{D}_n^+).

We see that with every one of our standard setups, the MD algorithm under appropriate conditions possesses dimension independent (or nearly dimension independent) complexity bound and, moreover, is nearly optimal in the sense of Information-based complexity theory, provided that the dimension is large.

Why the standard setups? “The contribution” of $\omega(\cdot)$ to the performance estimate (5.3.26) is in the factor Ω ; the less it is, the better. In principle, given X and $\|\cdot\|$, we could play with $\omega(\cdot)$ to minimize Ω . The standard setups are given by a kind of such optimization for the cases when X is the ball and $\|\cdot\| = \|\cdot\|_2$ (“the ball case”), when X is the simplex and $\|\cdot\| = \|\cdot\|_1$ (“the entropy case”), and when X is the ℓ_1/ℓ_2 ball, or the nuclear norm ball, and the norm $\|\cdot\|$ is the block ℓ_1 norm, respectively, the nuclear norm. We did not try to solve the arising variational problems exactly; however, it can be proved in all three cases that the value of Ω we have reached (i.e., $O(1)$ in the ball case, $O(\sqrt{\ln(n)})$ in the simplex case and the ℓ_1/ℓ_2 case, and $O(\sqrt{\ln(m)})$ in the nuclear norm case) cannot be reduced by more than an absolute constant factor.

Adjusting to problem’s geometry. A natural question is: When solving a particular problem (CP), which version of the Mirror Descent method to use? A “theory-based” answer could be “look at the complexity estimates of various versions of MD and run the one with the best complexity.” This “recommendation” usually is of no actual value, since the complexity bounds involve Lipschitz constants of the objective w.r.t. appropriate norms, and these constants usually are not known in advance. We are about to demonstrate that there are cases where reasonable recommendations can be made based solely on the geometry of the domain X of the problem. As an instructive example, consider the case where X is the unit $\|\cdot\|_p$ -ball: $X = X^p = \{x \in \mathbf{R}^n : \|x\|_p \leq 1\}$, where $p \in \{1, 2, \infty\}$.

Note that the cases of $p = 1$ and especially $p = \infty$ are “quite practical.” Indeed, the first of them ($p = 1$) is, essentially, what goes on in problems of ℓ_1 minimization (section 1.3.1); the second ($p = \infty$) is perhaps the most typical case (minimization under box constraints). The case of $p = 2$ seems to be more of academic nature.

The question we address is: which setup — Ball or Entropy one — to choose⁵.

First of all, we should explain how to process the domains in question via Entropy setup — the latter requires X to be a part of the full-dimensional standard simplex Δ_n^+ , which is *not* the case for the domains X we are interested in now. This issue can be easily resolved, specifically, as follows: setting

$$x_\rho[u, v] = \rho(u - v), \quad [u; v] \in \mathbf{R}^n \times \mathbf{R}^n,$$

we get a mapping from \mathbf{R}^{2n} onto \mathbf{R}^n such that the image of Δ_{2n}^+ under this mapping covers X^p , provided that $\rho = \rho_p = n^{\gamma_p}$, where $\gamma_1 = 0$, $\gamma_2 = 1/2$, $\gamma_\infty = 1$. It follows that in order to apply

⁵of course, we could look for more options as well, but as a matter of fact, in the case in question no more attractive options are known.

to (CP) the MD with Entropy setup, it suffices to rewrite the problem as

$$\min_{[u;v] \in Y^p} g_p(u, v) \equiv f(\rho_p(u - v)), \quad [Y_p = [u; v] \in \Delta_{2n}^+ : \rho_p[u - v] \in X_p]$$

the domain Y^p of the problem being a part of Δ_{2n}^+ .

The complexity bounds of the MD algorithm with both Ball and Entropy setups are of the form “ $N(\epsilon) = C\epsilon^{-2}$,” where the factor C is independent of ϵ , although depends on the problem at hand and on the setup we use. In order to understand which setup is more preferable, it suffices to look at the factors C only. These factors, up to $O(1)$ factors, are listed in following table (where $L_1(f)$ and $L_2(f)$ stand for the Lipschitz constants of f w.r.t. $\|\cdot\|_1$ - and $\|\cdot\|_2$ -norms):

Setup	$X = X^1$	$X = X^2$	$X = X^\infty$
Ball	$L_2^2(f)$	$L_2^2(f)$	$nL_2^2(f)$
Entropy	$L_1^2(f) \ln(n)$	$L_1^2(f)n \ln(n)$	$L_1^2(f)n^2 \ln(n)$

Now note that for $0 \neq x \in \mathbf{R}^n$ we have $1 \leq \frac{\|x\|_1}{\|x\|_2} \leq \sqrt{n}$, whence

$$1 \geq \frac{L_1^2(f)}{L_2^2(f)} \geq \frac{1}{n}.$$

With this in mind, the data from the above table suggest that *in the case of $p = 1$ (X is the ℓ_1 ball X^1), the Entropy setup is more preferable; in all remaining cases under consideration, the preferable setup if the Ball one.*

Indeed, when $X = X^1$, the Entropy setup can lose to the Ball one at most by the “small” factor $\ln(n)$ and can gain by factor as large as $n/\ln(n)$; these two options, roughly speaking, correspond to the cases when subgradients of f have just $O(1)$ significant entries (loss), or, on the contrary, have all entries of the same order of magnitude (gain). Since the potential loss is small, and potential gain can be quite significant, the Entropy setup in the case in question seems to be more preferable than the Ball one. In contrast to this, in the cases of $X = X^2$ and $X = X^\infty$ even the maximal potential gain associated with the Entropy setup (the ratio of $L_1^2(f)/L_2^2(f)$ as small as $1/n$) does not lead to an overall gain in the complexity, this is why in these cases we would advice to use the Ball setup.

5.3.5 Mirror Descent and Online Regret Minimization

5.3.5.1 Online regret minimization: what is it?

In the usual setting, the purpose of a black box oriented optimization algorithm as applied to a minimization problem

$$\text{Opt} = \min_{x \in X} f(x)$$

is to find an approximate solution of a desired accuracy ϵ via calls to an oracle (say, First Order one) representing f . What we are interested in, is the number N of steps (calls to the oracle) needed to build an ϵ -solution, and the overall computational effort in executing these N steps. What we ignore completely, is how “good,” in terms of f , are the search points $x_t \in X$, $1 \leq t \leq N$ – all we want is “to learn” f to the extent allowing to point out an ϵ -minimizer, and the faster is our learning process, the better, no matter how “expensive,” in terms of their non-optimality, are the search points. This traditional approach corresponds to the situation

where our optimization is *off line*: when processing the problem numerically, calling the oracle at a search point does not incur any actual losses. What indeed is used “in reality,” is the computed near-optimal solution, be it an engineering design, a reconstructed image, or a near-optimal portfolio selection. We are interested in losses incurred by this solution only, with no care of how good “in reality” would be the search points generated in course of the optimization process.

In *online* optimization, problem’s setting is different: we run an algorithm on the fly, in “real time,” and when calling the oracle at t -th step, the query point being x_t , incur “real loss,” in the simplest setting equal to $f(x_t)$. What we are interested in, is to make as close to Opt as possible our “properly averaged” loss, in the simplest case – the average loss

$$L_N = \frac{1}{N} \sum_{t=1}^N f(x_t).$$

The “horizon” N here can be fixed, in which case we want to make L_N as small as possible, or can vary, in which case we want to enforce L_N to approach Opt as fast as possible as N grows.

In fact, in online regret minimization it is not even assumed that the objective f remains constant as the time evolves; instead, it is assumed that we are dealing with a sequence of objectives f_t , $t = 1, 2, \dots$, selected “by nature” from a known in advance family \mathcal{F} of functions on a known in advance set $X \subset \mathbf{R}^n$. At a step $t = 1, 2, \dots$, we select a point $x_t \in X$, and “an oracle” provides us with certain information on f_t at x_t and charges us with “toll” $f_t(x_t)$. It should be stressed that in this setting, *no assumptions on how the nature selects the sequence $\{f_t(\cdot) \in \mathcal{F} : t \geq 1\}$ are made*. Now our “ideal goal” would be to find a *non-anticipative* policy (rules for generating x_t based solely on the information acquired from the oracle at the first $t-1$ steps of the process) which enforces our average toll

$$L_N = \frac{1}{N} \sum_{t=1}^N f_t(x_t) \tag{5.3.40}$$

to approach, as $N \rightarrow \infty$, the “ideal” average toll

$$L_N^* = \frac{1}{N} \sum_{t=1}^N \min_{x \in X} f_t(x), \tag{5.3.41}$$

whatever be the sequence $\{f_t(\cdot) \in \mathcal{F} : t \geq 1\}$ selected by the nature. It is absolutely clear that aside of trivial cases (like all functions from \mathcal{F} share common minimizer), this ideal goal cannot be achieved. Indeed, with $L_N - L_N^* \rightarrow 0$, $n \rightarrow \infty$, the average, over time horizon N , of (always nonnegative) non-optimalities $f_t(x_t) - \min_{x \in X} f_t(x)$ should go to 0 as $N \rightarrow \infty$, implying that for large N , “near all” of these non-optimalities should be “near-zero.” The latter, with the nature allowed to select $f_t \in \mathcal{F}$ in completely unpredictable manner, clearly cannot be guaranteed, whatever be non-anticipative policy used to generate x_1, x_2, \dots

To make the online setting with “abruptly varying” f_t ’s meaningful, people compare L_N not with its “utopian” lower bound L_N^* (which can be achieved only by a clairvoyant knowing in advance the trajectory f_1, f_2, \dots), but with a less utopian bound, most notably, with the quantity

$$\widehat{L}_N = \min_{x \in X} \frac{1}{N} \sum_{t=1}^N f_t(x).$$

Thus, we compare the average, on horizon N , toll of a “capable to move” (capable to select $x_t \in X$ in a whatever *non-anticipative* fashion) “mortal” which cannot predict the future with the best average toll of a clairvoyant who knows in advance the trajectory f_1, f_2, \dots selected by the nature, but is unable to move (must use the same action x during all “rounds” $t = 1, \dots, N$). We arrive at the *online regret minimization problem* where we are interested in a non-anticipative policy of selecting $x_t \in X$ which enforces the associated *regret*

$$\Delta_N = \sup_{f_t \in \mathcal{F}, 1 \leq t \leq N} \left[\frac{1}{N} \sum_{t=1}^N f_t(x_t) - \min_{x \in X} \frac{1}{N} \sum_{t=1}^N f_t(x) \right] \quad (5.3.42)$$

to go to 0, the faster the better, as $N \rightarrow \infty$, whatever be the sequence $\{f_t \in \mathcal{F} : t \geq 1\}$ selected by the nature.

Remark. Whether it is “fair” to compare a “mortal capable to move” with “a handicapped clairvoyant,” this question is beyond the scope of optimization theory, and answer here clearly depends on the application we are interested in. It suffices to say that there are meaningful applications, e.g., in Machine Learning, where online regret minimization makes sense and that even the case where the nature is restricted to select only stationary sequences $f_t(\cdot) \equiv f(\cdot) \in \mathcal{F}$, $t = 1, 2, \dots$ (no difficulty with “fairness” at all), making Δ_N small models the meaningful situation when we “learn” the objective f online.

5.3.5.2 Online regret minimization via Mirror Descent, deterministic case

Note that the problem of online regret minimization admits various settings, depending on what are our assumptions on X and \mathcal{F} , and what is the information on f_t revealed by the oracle at the search point x_t . In this section, we focus on the simple case where

1. X is a convex compact set in Euclidean space E , and (X, E) is equipped with proximal setup – a norm $\|\cdot\|$ on E and a compatible with this norm DGF $\omega(\cdot)$, which we assume to be continuously differentiable on X ;
2. \mathcal{F} is the family of all convex and Lipschitz continuous, with a given constant L w.r.t. $\|\cdot\|$, functions on X ;
3. the information we get in round t , our search point being x_t , and “nature’s selection” being $f_t(\cdot) \in \mathcal{F}$, is the first order information $f_t(x_t), f'_t(x_t)$ on f_t at x_t , with the $\|\cdot\|_*$ -norm of the taken at x_t subgradient $f'_t(x_t)$ of $f_t(\cdot)$ not exceeding L .

Our goal is to demonstrate that Mirror Descent algorithm, in the form presented in section 5.3.4.5, is well-suited for online regret minimization. Specifically, consider algorithm as follows:

1. *Initialization:* Select stepsizes $\{\gamma_t > 0\}_{t \geq 1}$ with γ_t nonincreasing in t . Select $x_1 \in X$.
2. *Step* $t = 1, 2, \dots$. Given x_t and a reported by the oracle subgradient $f'_t(x_t)$, of $\|\cdot\|_*$ -norm not exceeding L , of f_t at x_t , set

$$x_{t+1} = \text{Prox}_{x_t}(\gamma_t f'_t(x_t)) := \operatorname{argmin}_{x \in X} [\langle \gamma_t f'_t(x_t) - \omega'(x_t), x \rangle + \omega(x)]$$

and pass to step $t + 1$.

Convergence analysis. We have the following straightforward modification of Theorem 5.3.2:

Theorem 5.3.3 *In the notation and under assumptions of this section, whatever be a sequence $\{f_t \in \mathcal{F} : t \geq 1\}$ selected by the nature, for the trajectory $\{x_t : t \geq 1\}$ of search points generated by the above algorithm and for a whatever sequence of weights $\{\lambda_t > 0 : t \geq 1\}$ such that λ_t/γ_t is nondecreasing in t , it holds for all $N = 1, 2, \dots$:*

$$\frac{1}{\Lambda_N} \sum_{t=1}^N \lambda_t f_t(x_t) - \min_{u \in X} \frac{1}{\Lambda_N} \sum_{t=1}^N \lambda_t f_t(u) \leq \frac{\frac{\lambda_N}{\gamma_N} \Omega_*^2 + \sum_{t=1}^N \lambda_t \gamma_t L^2}{2\Lambda_N} \quad (5.3.43)$$

where

$$\Lambda_N = \sum_{t=1}^N \lambda_t, \quad \Omega_* = \sup_{x, y \in X} \sqrt{2V_x(y)} \equiv \sup_{x, y \in X} \sqrt{2[\omega(y) - \omega(x) - \langle \omega'(x), y - x \rangle]}.$$

Postponing for a moment the proof, let us look at the consequences.

Consequences. Let us choose

$$\gamma_t = \frac{\Omega_* \sqrt{1 + \alpha}}{L \sqrt{t}}, \quad \lambda_t = t^\alpha,$$

with $\alpha \geq 0$. This choice meets the requirements from the description of our algorithm and the premise of Theorem 5.3.3, and results in $\Lambda_N \geq O(1)(1 + \alpha)^{-1}N^{\alpha+1}$ and $\sum_{t=1}^N \lambda_t \gamma_t L^2 \leq O(1)(1 + \alpha)^{-1}N^{\alpha+1/2}(\Omega_* L)$, provided $N \geq 1 + \alpha$. As a result, (5.3.43) becomes

$$\begin{aligned} & \forall (f_t \in \mathcal{F}, t = 1, 2, \dots, N \geq 1 + \alpha) : \\ & \frac{1}{\Lambda_N} \sum_{t=1}^N \lambda_t f_t(x_t) - \min_{u \in X} \frac{1}{\Lambda_N} \sum_{t=1}^N \lambda_t f_t(u) \leq O(1) \frac{\sqrt{1+\alpha} \Omega_* L}{\sqrt{N}}. \end{aligned} \quad (5.3.44)$$

We arrive at an $O(1/\sqrt{N})$ upper bound on a kind of regret (“kind of” stems from the fact that the averaging in the left hand side uses weights proportional to λ_t rather than equal to each other). Setting $\alpha = 0$, we get a bound on the “true” N -step regret. Note that the policy of generating the points x_1, x_2, \dots is completely independent of α , so that the bound (5.3.44) contains a “spectrum,” parameterized by $\alpha \geq 0$, of regret-like bounds on a fixed decision-making policy.

Proof of Theorem 5.3.3. With our current algorithm, relation (5.3.23) becomes

$$\forall (t \geq 1, u \in X) : \gamma_t \langle f'_t(x_\tau), x_t - u \rangle \leq V_{x_t}(u) - V_{x_{t+1}}(u) + \frac{1}{2} \|\gamma_t f'_t(x_t)\|_*^2 \quad (5.3.45)$$

(replace f with f_t in the derivation in item 1⁰ of the proof of Theorem 5.3.1). Now let us proceed exactly as in the proof of Theorem 5.3.2: multiply both sides in (5.3.45) by λ_t/γ_t and sum up the resulting inequalities over $t = 1, \dots, N$, arriving at

$$\begin{aligned} & \forall (u \in X) : \\ & \sum_{t=1}^N \lambda_t \langle f'_t(x_t), x_t - u \rangle \leq \sum_{t=1}^N \frac{\lambda_t}{\gamma_t} [V_{x_t}(u) - V_{x_{t+1}}(u)] + \frac{1}{2} \sum_{t=1}^N \lambda_t \gamma_t \|f'_t(x_t)\|_*^2 \\ & = \underbrace{\frac{\lambda_1}{\gamma_1} V_{x_1}(u) + \left[\underbrace{\frac{\lambda_2}{\gamma_2} - \frac{\lambda_1}{\gamma_1}}_{\geq 0} \right] V_{x_2}(u) + \left[\underbrace{\frac{\lambda_3}{\gamma_3} - \frac{\lambda_2}{\gamma_2}}_{\geq 0} \right] V_{x_3}(u) + \dots + \left[\underbrace{\frac{\lambda_N}{\gamma_N} - \frac{\lambda_{N-1}}{\gamma_{N-1}}}_{\geq 0} \right] V_{x_N}(u)}_{\geq 0} \\ & - \frac{\lambda_N}{\gamma_N} V_{x_{N+1}}(u) + \frac{1}{2} \sum_{t=1}^N \lambda_t \gamma_t \|f'_t(x_t)\|_*^2 \leq \frac{\lambda_N}{\gamma_N} \frac{\Omega_*^2}{2} + \frac{1}{2} \sum_{t=1}^N \lambda_t \gamma_t \|f'_t(x_t)\|_*^2, \end{aligned}$$

where the concluding inequality is due to $0 \leq V_{x_t}(u) \leq \frac{\Omega_*^2}{2}$ for all t . As a result, taking into account that $\|f'_t(x_t)\|_* \leq L$, we get

$$\max_{u \in X} \Lambda_N^{-1} \sum_{t=1}^N \lambda_t \langle f'_t(x_t), x_t - u \rangle \leq \frac{\frac{\lambda_N}{\gamma_N} \Omega_*^2 + \sum_{t=1}^N \lambda_t \gamma_t L^2}{2\Lambda_N} \quad (5.3.46)$$

$$[\Lambda_N = \sum_{t=1}^N \lambda_t].$$

On the other hand, by convexity of f_t we have for every $u \in X$:

$$f_t(x_t) - f_t(u) \leq \langle f'_t(x_t), x_t - u \rangle,$$

and thus (5.3.46) implies that

$$\frac{1}{\Lambda_N} \sum_{t=1}^N \lambda_t f_t(x_t) - \min_{u \in X} \frac{1}{\Lambda_N} \sum_{t=1}^N \lambda_t f_t(u) \leq \frac{\frac{\lambda_N}{\gamma_N} \Omega_*^2 + \sum_{t=1}^N \lambda_t \gamma_t L^2}{2\Lambda_N}. \quad \square$$

5.3.6 Mirror Descent for Saddle Point problems

5.3.6.1 Convex-Concave Saddle Point problem

An important generalization of a convex optimization problem (CP) is a *convex-concave saddle point problem*

$$\min_{x \in X} \max_{y \in Y} \phi(x, y) \quad (\text{SP})$$

where

- $X \subset \mathbf{R}^{n_x}$, $Y \subset \mathbf{R}^{n_y}$ are nonempty convex compact subsets in the respective Euclidean spaces (which at this point we w.l.o.g. identified with a pair of \mathbf{R}^n 's),
- $\phi(x, y)$ is a Lipschitz continuous real-valued *cost function* on $Z = X \times Y$ which is convex in $x \in X$, $y \in Y$ being fixed, and is concave in $y \in Y$, $x \in X$ being fixed.

Basic facts on saddle points. (SP) gives rise to a pair of convex optimization problems

$$\begin{aligned} \text{Opt}(P) &= \min_{x \in X} \bar{\phi}(x) := \max_{y \in Y} \phi(x, y) \quad (\text{P}) \\ \text{Opt}(D) &= \max_{y \in Y} \underline{\phi}(y) := \min_{x \in X} \phi(x, y) \quad (\text{D}) \end{aligned}$$

(the *primal* problem (P) requires to minimize a convex function over a convex compact set, the *dual* problem (D) requires to maximize a concave function over another convex compact set). The problems are *dual to each other*, meaning that $\text{Opt}(P) = \text{Opt}(D)$; this is called *strong duality*⁶. Note that strong duality is based upon convexity-concavity of ϕ and convexity of X, Y (and to a slightly lesser extent – on the continuity of ϕ on $Z = X \times Y$ and compactness of the latter domain), while the *weak duality* $\text{Opt}(P) \geq \text{Opt}(D)$ is a completely general (and simple to verify) phenomenon.

In our case of convex-concave Lipschitz continuous ϕ the induced objectives in the primal and the dual problem are Lipschitz continuous as well. Since the domains of these problems are compact, the problems are solvable. Denoting by X_* , Y_* the corresponding optimal sets, the

⁶For missing proofs of all claims in this section, see section D.4.

set $Z_* = X_* \times Y_*$ admits simple characterization in terms of ϕ : Z_* is exactly the set of saddle points of ϕ on $Z = X \times Y$, that is, the points $(x_*, y_*) \in X \times Y$ such that

$$\phi(x, y_*) \geq \phi(x_*, y_*) \geq \phi(x_*, y) \quad \forall (x, y) \in Z = X \times Y$$

(in words: with y set to y_* , ϕ attains its minimum over $x \in X$ at x_* , and with x set to x_* , ϕ attains its maximum over $y \in Y$ at $y = y_*$). These points are considered as exact solutions to the saddle point problem in question. The common optimal value of (P) and (D) is nothing but the value of ϕ at a saddle point:

$$\text{Opt}(P) = \text{Opt}(D) = \phi(x_*, y_*) \quad \forall (x_*, y_*) \in Z_*.$$

The standard interpretation of a saddle point problem is a *game*, where the first player chooses his action x from the set X , his adversary – the second player – chooses his action y in Y , and with a choice (x, y) of the players, the first pays the second the sum $\phi(x, y)$. Such a game is called *antagonistic* (since the interests of the players are completely opposite to each other), or a *zero sum game* (since the sum $\phi(x, y) + [-\phi(x, y)]$ of losses of the players is identically zero). In terms of the game, saddle points (x_*, y_*) are exactly the equilibria: if the first player sticks to the choice x_* , the adversary has no incentive to move away from the choice y_* , since such a move cannot increase his profit, and similarly for the first player, provided that the second one sticks to the choice y_* .

It is convenient to characterize the accuracy of a candidate $(x, y) \in X \times Y$ to the role of an approximate saddle point by the “saddle point residual”

$$\begin{aligned} \epsilon_{\text{sad}}(x, y) &:= \max_{y' \in Y} \phi(x, y') - \min_{x' \in X} \phi(x', y) = \bar{\phi}(x) - \underline{\phi}(y) \\ &= [\bar{\phi}(x) - \text{Opt}(P)] + [\text{Opt}(D) - \underline{\phi}(y)], \end{aligned}$$

where the concluding equality is due to strong duality. We see that the saddle point inaccuracy $\epsilon_{\text{sad}}(x, y)$ is just the sum of non-optimalities, in terms of the respective objectives, of x as a solution to the primal, and y as a solution to the dual problem; in particular, the inaccuracy is nonnegative and is zero if and only if (x, y) is a saddle point of ϕ .

Vector field associated with (SP). From now on, we focus on problem (SP) with Lipschitz continuous convex-concave cost function $\phi : X \times Y \rightarrow \mathbf{R}$ and convex compact X, Y and assume that ϕ is represented by a *First Order oracle* as follows: given on input a point $z = (x, y) \in Z := X \times Y$, the oracle returns a subgradient $F_x(z)$ of the convex function $\phi(\cdot, y) : X \rightarrow \mathbf{R}$ at the point x , and a subgradient $F_y(x, y)$ of the convex function $-\phi(x, \cdot) : Y \rightarrow \mathbf{R}$ at the point Y . It is convenient to arrange F_x, F_y in a single vector

$$F(z) = [F_x(z); F_y(z)] \in \mathbf{R}^{n_x} \times \mathbf{R}^{n_y}. \quad (5.3.47)$$

In addition, we assume that the vector field $\{F(z) : z \in Z\}$ is bounded (cf. (5.3.1)).

5.3.6.2 Saddle point MD algorithm

The setup for the algorithm is a usual proximal setup, but with the set $Z = X \times Y$ in the role of X and the embedding space $\mathbf{R}^{n_x} \times \mathbf{R}^{n_y}$ of Z , in the role of \mathbf{R}^n , so that the setup is given by a norm $\|\cdot\|$ on $\mathbf{R}^{n_x} \times \mathbf{R}^{n_y}$ and a DGF $\omega(\cdot) : Z \rightarrow \mathbf{R}$. We change the notation accordingly; e.g., Z° is the set of those $z \in Z$ where $\omega(\cdot)$ admits a subgradient, etc.

The MD algorithm for (SP) is given by the recurrence

$$\begin{aligned} z_1 := (x_1, y_1) &= z_\omega := \operatorname{argmin}_{z \in Z} \omega(z), \\ z_{t+1} := (x_{t+1}, y_{t+1}) &= \operatorname{Prox}_{z_t}(\gamma_t F(z_t)), \\ \operatorname{Prox}_z(\xi) &= \operatorname{argmin}_{w \in Z} [V_z(w) + \langle \xi, w \rangle], \\ V_z(w) &= \omega(w) - \langle \omega'(z), w - z \rangle - \omega(z) \geq \frac{1}{2} \|w - z\|^2, \\ z^t &= \left[\sum_{\tau=1}^t \gamma_\tau \right]^{-1} \sum_{\tau=1}^t \gamma_\tau z_\tau, \end{aligned} \quad (5.3.48)$$

where $\gamma_t > 0$ are stepsizes, and z^t are approximate solutions built in course of t steps.

The saddle point analogy of Theorem 5.3.1 is as follows:

Theorem 5.3.4 *For every $\tau \geq 1$ and every $w = (u, v) \in Z$, one has*

$$\gamma_\tau \langle F(z_\tau), z_\tau - w \rangle \leq V_{z_\tau}(w) - V_{z_{\tau+1}}(w) + \frac{1}{2} \|\gamma_\tau F(z_\tau)\|_*^2. \quad (5.3.49)$$

As a result, for every $t \geq 1$ one has

$$\epsilon_{\text{sad}}(z^t) \leq \frac{\frac{\Omega^2}{2} + \frac{1}{2} \sum_{\tau=1}^t \gamma_\tau^2 \|F(z_\tau)\|_*^2}{\sum_{\tau=1}^t \gamma_\tau}. \quad (5.3.50)$$

In particular, given a number T of iterations we intend to perform and setting

$$\gamma_t = \frac{\Omega}{\|F(z_t)\|_* \sqrt{T}}, \quad 1 \leq t \leq T, \quad (5.3.51)$$

we ensure that

$$\epsilon_{\text{sad}}(z^T) \leq \Omega \frac{\sqrt{T}}{\sum_{t=1}^T \frac{1}{\|F(x_t)\|_*}} \leq \frac{\Omega \max_{t \leq T} \|F(x_t)\|_*}{\sqrt{T}} \leq \frac{\Omega L_{\|\cdot\|}(F)}{\sqrt{T}}, \quad L_{\|\cdot\|}(F) = \sup_{z \in Z} \|F(z)\|_*. \quad (5.3.52)$$

Proof. 1⁰. (5.3.49) is proved exactly as its ‘‘minimization counterpart’’ (5.3.23), see item 1⁰ of the proof of Theorem 5.3.1; on a closest inspection, the reasoning there is absolutely independent of where the vectors ξ_t in the recurrence $z_{t+1} = \operatorname{Prox}_{z_t}(\xi_t)$ come from.

2⁰. Summing up relations (5.3.49) over $\tau = 1, \dots, t$, plugging in $F = [F_x; F_y]$, dividing both sides of the resulting inequality by $\Gamma_t = \sum_{\tau=1}^t \gamma_\tau$ and setting $\nu_\tau = \gamma_\tau / \Gamma_t$, we get

$$\sum_{\tau=1}^T \nu_\tau [\langle F_x(z_\tau), x_\tau - u \rangle + \langle F_y(z_\tau), y_\tau - v \rangle] \leq Q := \Gamma_t^{-1} \left[\frac{\Omega^2}{2} + \frac{1}{2} \sum_{\tau=1}^t \gamma_\tau^2 \|F(z_\tau)\|_*^2 \right]; \quad (!)$$

this inequality holds true for all $(u, v) \in X \times Y$. Now, recalling the origin of F_x and F_y , we have

$$\langle F_x(x_t, y_t), x_t - u \rangle \geq \phi(x_t, y_t) - \phi(u, y_t), \quad \langle F_y(x_t, y_t), y_t - v \rangle \geq \phi(x_t, v) - \phi(x_t, y_t),$$

so that (!) implies that

$$\sum_{\tau=1}^t \nu_\tau [\phi(x_t, v) - \phi(u, y_t)] \leq Q := \Gamma_t^{-1} \left[\frac{\Omega^2}{2} + \frac{1}{2} \sum_{\tau=1}^t \gamma_\tau^2 \|F(z_\tau)\|_*^2 \right],$$

which, by convexity-concavity of ϕ , yields

$$[\phi(x^t, v) - \phi(u, y^t)] \leq Q := \Gamma_t^{-1} \left[\frac{\Omega^2}{2} + \frac{1}{2} \sum_{\tau=1}^t \gamma_\tau^2 \|F(z_\tau)\|_*^2 \right].$$

Now, the right hand side of the latter inequality is independent of $(u, v) \in Z$; taking maximum over $(u, v) \in Z$ in the left hand side of the inequality, we get (5.3.50). This inequality clearly implies (5.3.52). \square

In our course, the role of Theorem 5.3.4 (which is quite useful by its own right) is to be a “draft version” of a much more powerful Theorem 5.5.4, and we postpone illustrations till the latter Theorem is in place.

5.3.6.3 Refinement

Similarly to the minimization case, the complexity analysis of saddle point MD can be refined. Specifically, assume that for properly selected Ω_* , $0 < \Omega_* < \infty$, we have

$$\sup_{z \in Z^o, u \in Z} V_z(u) \leq \frac{\Omega_*^2}{2}. \quad (5.3.53)$$

(cf. (5.3.27)). Then Theorem 5.3.4 admits a refinement as follows:

Theorem 5.3.5 *Let (5.3.53) take place, and let $\lambda_1, \lambda_2, \dots$ be positive reals such that (5.3.28) holds true, and let*

$$z^t = \frac{\sum_{\tau=1}^t \lambda_\tau z_\tau}{\sum_{\tau=1}^t \lambda_\tau}.$$

Then $z^t \in Z$, and

$$\epsilon_{\text{sad}}(z^t) \leq \frac{\frac{\lambda_t}{\gamma_t} \Omega_*^2 + \sum_{\tau=1}^t \lambda_t \gamma_\tau \|F(z_\tau)\|_*^2}{2 \sum_{\tau=1}^t \lambda_\tau}. \quad (5.3.54)$$

In particular, with the stepsizes

$$\gamma_t = \frac{\Omega_*}{\|F(z_t)\|_* \sqrt{t}}, \quad t = 1, 2, \dots \quad (5.3.55)$$

and with

$$\lambda_t = \frac{t^p}{\|F(z_t)\|_*}, \quad t = 1, 2, \dots \quad (5.3.56)$$

where $p > -1/2$, we ensure for $t = 1, 2, \dots$ that

$$\epsilon_{\text{sad}}(z^t) \leq C_p \frac{\Omega_* L_{\|\cdot\|}(F)}{\sqrt{t}}, \quad L_{\|\cdot\|}(F) = \sup_{z \in Z} \|F(z)\|_* \quad (5.3.57)$$

with C_p depending solely and continuously on $p > -1/2$.

Proof. The inclusion $z^t \in Z$ is evident. Processing (5.3.49) in exactly the same fashion as we have processed (5.3.23) when proving Theorem 5.3.2, we get the relation

$$\sup_{u \in Z} \Lambda_t^{-1} \sum_{\tau=1}^t \lambda_\tau \langle F(z_t), z_t - u \rangle \leq \frac{\frac{\lambda_t}{\gamma_t} \Omega_*^2 + \sum_{\tau=1}^t \lambda_\tau \gamma_\tau \|F(z_\tau)\|_*^2}{\Lambda_t}$$

$$\left[\Lambda_t = \sum_{\tau=1}^t \lambda_\tau \right]$$

Same as in the proof of Theorem 5.3.4, the left hand side in this inequality upper-bounds $\epsilon_{\text{sad}}(z^t)$, and we arrive at (5.3.54). The “In particular” part of Theorem is straightforwardly implied by (5.3.54). \square

5.3.7 Mirror Descent for Stochastic Minimization/Saddle Point problems

5.3.7.1 Stochastic Minimization/Saddle Point problems

Let we need to solve a saddle point problem

$$\min_{x \in X} \max_{y \in Y} \phi(x, y) \quad (\text{SP})$$

under the same assumptions as above (i.e., X, Y are convex compact sets in Euclidean spaces $\mathbf{R}^{n_x}, \mathbf{R}^{n_y}$, $\phi : Z = X \times Y \rightarrow \mathbf{R}$ is convex-concave and Lipschitz continuous), but now, instead of deterministic First Order oracle which reports the values of the associated with ϕ vector field F , see (5.3.47), we have at our disposal a *stochastic First Order oracle* – a routine which reports unbiased random estimates of F rather than the exact values of F . A convenient model of the oracle is as follows: at t -th call to the oracle, the query point being $z \in Z = X \times Y$, the oracle returns a vector $G = G(z, \xi_t)$ which is a deterministic function of z and of t -th realization of oracle’s noise ξ_t . We assume the noises ξ_1, ξ_2, \dots to be independent of each other and identically distributed: $\xi_t \sim P, t = 1, 2, \dots$. We always assume the oracle is unbiased:

$$\mathbf{E}_{\xi \sim P}\{G(z, \xi)\} =: F(z) = [F_x(z); F_y(z)] \in \partial_x \phi(z) \times \partial_y [-\phi(z)] \quad (5.3.58)$$

(thus, $F_x(x, y)$ is a subgradient of $\phi(\cdot, y)$ taken at the point x , while $F_y(x, y)$ is a subgradient of $-\phi(x, \cdot)$ taken at the point y). Besides this, we assume that the second moment of the stochastic subgradient is bounded uniformly in $z \in Z$:

$$\sup_{z \in Z} \mathbf{E}_{\xi \sim P}\{\|G(z, \xi)\|_2^2\} < \infty, \quad (5.3.59)$$

Note that when Y is a singleton, our setting recovers the problem of minimizing a Lipschitz continuous convex function over a compact convex set in the case when instead of the subgradients of the objective, unbiased random estimates of these subgradients are available.

5.3.7.2 Stochastic Saddle Point Mirror Descent algorithm

It turns out that the MD algorithm admits a stochastic version capable to solve (SP) in the outlined “stochastic environment.” The setup for the algorithm is a usual proximal setup — a norm $\|\cdot\|$ on the space $\mathbf{R}^{n_x} \times \mathbf{R}^{n_y}$ embedding X and a DGF $\omega(\cdot)$ for Z . The Stochastic Saddle Point Mirror Descent (SSPMD) algorithm is identical to its deterministic counterpart; it is the recurrence

$$z_1 = z_\omega, z_{t+1} = \text{Prox}_{z_t}(\gamma_t G(z_t, \xi_t)), z^t = \left[\sum_{\tau=1}^t \gamma_\tau \right]^{-1} \sum_{\tau=1}^t \gamma_\tau z_\tau, \quad (5.3.60)$$

where $\gamma_t > 0$ are deterministic stepsizes. This is nothing but the recurrence (5.3.48), with the only difference that now this recurrence “is fed” by the estimates $G(z_t, \xi_t)$ of the vectors $F(z_t)$ participating in (5.3.48). Surprisingly, the performance of SSPMD is, essentially, the same as the performance of its deterministic counterpart:

Theorem 5.3.6 One has

$$\begin{aligned} \mathbf{E}_{[\xi_1; \dots; \xi_t] \sim P \times \dots \times P} \{ \epsilon_{\text{sad}}(z^t) \} &\leq \frac{7}{2} \frac{\Omega^2 + \sigma^2 \sum_{\tau=1}^t \gamma_\tau^2}{\sum_{\tau=1}^t \gamma_\tau}. \\ \sigma^2 &= \sup_{z \in Z} \mathbf{E}_{\xi \sim P} \{ \|G(z, \xi)\|_*^2 \} \end{aligned} \quad (5.3.61)$$

In particular, given a number T of iterations we intend to perform and setting

$$\gamma_t = \frac{\Omega}{\sigma \sqrt{T}}, \quad 1 \leq t \leq T, \quad (5.3.62)$$

we ensure that

$$\mathbf{E}\{\epsilon_{\text{sad}}(z^T)\} \leq \frac{7\Omega\sigma}{\sqrt{T}}. \quad (5.3.63)$$

Proof. 1⁰. By exactly the same reason as in the deterministic case, we have for every $u \in Z$:

$$\sum_{\tau=1}^t \gamma_\tau \langle G(z_\tau, \xi_\tau), z_\tau - u \rangle \leq \frac{\Omega^2}{2} + \frac{1}{2} \sum_{\tau=1}^t \gamma_\tau^2 \|G(z_\tau, \xi_\tau)\|_*^2.$$

Setting

$$\Delta_\tau = G(z_\tau, \xi_\tau) - F(z_\tau),$$

we can rewrite the above relation as

$$\sum_{\tau=1}^t \gamma_\tau \langle F(z_\tau), z_\tau - u \rangle \leq \frac{\Omega^2}{2} + \frac{1}{2} \sum_{\tau=1}^t \gamma_\tau^2 \|G(z_\tau, \xi_\tau)\|_*^2 + \underbrace{\sum_{\tau=1}^t \gamma_\tau \langle \Delta_\tau, u - z_\tau \rangle}_{D(\xi^t, u)}, \quad (5.3.64)$$

where $\xi^t = [\xi_1; \dots; \xi_t]$. It follows that

$$\max_{u \in Z} \sum_{\tau=1}^t \gamma_\tau \langle F(z_\tau), z_\tau - u \rangle \leq \frac{\Omega^2}{2} + \frac{1}{2} \sum_{\tau=1}^t \gamma_\tau^2 \|G(z_\tau, \xi_\tau)\|_*^2 + \max_{u \in Z} D(\xi^t, u).$$

As we have seen when proving Theorem 5.3.4, the left hand side in the latter relation is $\geq [\sum_{\tau=1}^t \gamma_\tau] \epsilon_{\text{sad}}(z^t)$, so that

$$\left[\sum_{\tau=1}^t \gamma_\tau \right] \epsilon_{\text{sad}}(z^t) \leq \frac{\Omega^2}{2} + \frac{1}{2} \sum_{\tau=1}^t \gamma_\tau^2 \|G(z_\tau, \xi_\tau)\|_*^2 + \max_{u \in Z} D(\xi^t, u). \quad (5.3.65)$$

Now observe that by construction of the method z_τ is a deterministic function of $\xi^{\tau-1}$: $z_\tau = Z_\tau(\xi^{\tau-1})$, where Z_τ takes its values in Z . Due to this observation and to the fact that γ_τ are deterministic, taking expectations of both sides in (5.3.65) w.r.t. ξ^t we get

$$\left[\sum_{\tau=1}^t \gamma_\tau \right] \mathbf{E}\{\epsilon_{\text{sad}}(z^t)\} \leq \frac{\Omega^2}{2} + \frac{\sigma^2}{2} \sum_{\tau=1}^t \gamma_\tau^2 + \mathbf{E} \left\{ \max_{u \in Z} D(\xi^t, u) \right\}; \quad (5.3.66)$$

note that in this computation we bounded $\mathbf{E}\{\|G(z_\tau, \xi_\tau)\|_*^2\}$, $\tau \leq t$, as follows:

$$\mathbf{E}\left\{\|G(z_\tau, \xi_\tau)\|_*^2\right\} = \mathbf{E}_{\xi^t} \left\{ \|G(Z_\tau(\xi^{\tau-1}), \xi_\tau)\|_*^2 \right\} = \mathbf{E}_{\xi^{\tau-1}} \left\{ \underbrace{\mathbf{E}_{\xi_\tau} \left\{ \|G(Z_\tau(\xi^{\tau-1}), \xi_\tau)\|_*^2 \right\}}_{\leq \sigma^2} \right\} \leq \sigma^2.$$

2⁰. It remains to bound $\mathbf{E}\{\max_{u \in Z} D(\xi^t, u)\}$, which is done in the following two lemmas:

Lemma 5.3.1 For $1 \leq \tau \leq t$, let v_τ be a deterministic vector-valued function of $\xi^{\tau-1}$: $v_\tau = V_\tau(\xi^{\tau-1})$ taking values in Z . Then

$$\mathbf{E} \left\{ \left| \underbrace{\sum_{\tau=1}^t \gamma_\tau \langle \Delta_\tau, v_\tau - z_\omega \rangle}_{\psi_\tau} \right| \right\} \leq 2\sigma\Omega \sqrt{\sum_{\tau=1}^t \gamma_\tau^2}. \quad (5.3.67)$$

Proof. Setting $S_0 = 0$, $S_\tau = \sum_{s=1}^\tau \gamma_s \langle \Delta_s, v_s - z_\omega \rangle$, we have

$$\mathbf{E} \{ S_{\tau+1}^2 \} = \mathbf{E} \{ S_{\tau-1}^2 + 2S_{\tau-1}\psi_\tau + \psi_\tau^2 \} \leq \mathbf{E} \{ S_{\tau-1}^2 \} + 4\sigma^2\Omega^2\gamma_\tau^2, \quad (5.3.68)$$

where the concluding \leq comes from the following considerations:

- a) By construction, $S_{\tau-1}$ depends solely on $\xi^{\tau-1}$, while $\psi_\tau = \gamma_\tau \langle G(Z_\tau(\xi^{\tau-1}), \xi_\tau) - F(Z_\tau(\xi^{\tau-1})), V_\tau(\xi^{\tau-1}) \rangle$. Invoking (5.3.58), the expectation of ψ_τ w.r.t. ξ_τ is zero for all $\xi^{\tau-1}$, whence the expectation of the product of $S_{\tau-1}$ (this quantity depends solely of $\xi^{\tau-1}$) and ψ_τ is zero.
- b) Since $\mathbf{E}_{\xi \sim P} \{ G(z, \xi) \} = F(z)$ for all $z \in Z$ and due to the origin of σ , we have $\|F(z)\|_*^2 \leq \sigma^2$ for all $z \in Z$, whence $\mathbf{E}_{\xi \sim P} \{ \|G(z, \xi) - F(z)\|_*^2 \} \leq 4\sigma^2$, whence $\mathbf{E} \{ \|\Delta_\tau\|_*^2 \} \leq 4\sigma^2$ as well due to $\Delta_\tau = G(Z_\tau(\xi^{\tau-1}), \xi_\tau) - F(Z_\tau(\xi^{\tau-1}))$. Since v_τ takes its values in Z , we have $\|z_\omega - v_\tau\|^2 \leq \Omega^2$, see (5.3.7), whence $|\psi_\tau| \leq \gamma_\tau \Omega \|\Delta_\tau\|_*$, and therefore $\mathbf{E} \{ \psi_\tau^2 \} \leq 4\sigma^2\Omega^2\gamma_\tau^2$.

From (5.3.68) it follows that $\mathbf{E} \{ S_t^2 \} \leq 4 \sum_{\tau=1}^t \sigma^2\Omega^2\gamma_\tau^2$, whence

$$\mathbf{E} \{ |S_t| \} \leq 2 \sqrt{\sum_{\tau=1}^t \sigma^2\gamma_\tau^2\Omega^2}. \quad \square$$

Lemma 5.3.2 We have

$$\mathbf{E} \left\{ \max_{u \in Z} D(\xi^t, u) \right\} \leq 6\sigma\Omega \sqrt{\sum_{\tau=1}^t \gamma_\tau^2}. \quad (5.3.69)$$

Proof. Let $\rho > 0$. Consider an auxiliary recurrence as follows:

$$v_1 = z_\omega; v_{\tau+1} = \text{Prox}_{v_\tau}(-\rho\gamma_\tau\Delta_\tau).$$

Recalling that Δ_τ is a deterministic function of $\xi^{\tau-1}$, we see that v_τ is a deterministic function of $\xi^{\tau-1}$ taking values in Z . Same as at the first step of the proof of Theorem 5.3.4, the recurrence specifying v_τ implies that for every $u \in Z$ it holds

$$\sum_{\tau=1}^t \rho\gamma_\tau \langle -\Delta_\tau, v_\tau - u \rangle \leq \frac{\Omega^2}{2} + \frac{\rho^2}{2} \sum_{\tau=1}^t \gamma_\tau^2 \|\Delta_\tau\|_*^2,$$

or, equivalently,

$$\sum_{\tau=1}^t \gamma_\tau \langle -\Delta_\tau, v_\tau - u \rangle \leq \frac{\Omega^2}{2\rho} + \frac{\rho}{2} \sum_{\tau=1}^t \gamma_\tau^2 \|\Delta_\tau\|_*^2,$$

whence

$$\begin{aligned} D(\xi^t, u) &= \sum_{\tau=1}^t \gamma_\tau \langle -\Delta_\tau, z_\tau - u \rangle \leq \sum_{\tau=1}^t \gamma_\tau \langle -\Delta_\tau, v_\tau - u \rangle + \sum_{\tau=1}^t \gamma_\tau \langle -\Delta_\tau, z_\tau - v_\tau \rangle \\ &\leq Q_\rho(\xi^t) := \frac{\Omega^2}{2\rho} + \frac{\rho}{2} \sum_{\tau=1}^t \gamma_\tau^2 \|\Delta_\tau\|_*^2 + \sum_{\tau=1}^t \gamma_\tau \langle \Delta_\tau, v_\tau - z_\omega \rangle + \sum_{\tau=1}^t \gamma_\tau \langle \Delta_\tau, z_\omega - z_\tau \rangle \end{aligned}$$

Note that $Q_\rho(\xi^t)$ indeed depends solely on ξ^t , so that the resulting inequality says that

$$\max_{u \in Z} D(\xi^t, u) \leq Q_\rho(\xi^t) \Rightarrow \mathbf{E} \left\{ \max_{u \in Z} D(\xi^t, u) \right\} \leq \mathbf{E}\{Q_\rho(\xi^t)\}.$$

Further, by construction both v_τ and z_τ are deterministic functions of $\xi^{\tau-1}$ taking values in Z , and invoking Lemma 5.3.1 and taking into account that $\mathbf{E}\{\|\Delta_\tau\|_*^2\} \leq 4\sigma^2$, we get

$$\mathbf{E}\{Q_\rho(\xi^t)\} \leq \frac{\Omega^2}{2\rho} + 2\rho\sigma^2 \sum_{\tau=1}^t \gamma_\tau^2 + 4\sigma\Omega \sqrt{\sum_{\tau=1}^t \gamma_\tau^2},$$

whence

$$\mathbf{E} \left\{ \max_{u \in Z} D(\xi^t, u) \right\} \leq \frac{\Omega^2}{2\rho} + 2\rho\sigma^2 \sum_{\tau=1}^t \gamma_\tau^2 + 4\sigma\Omega \sqrt{\sum_{\tau=1}^t \gamma_\tau^2}.$$

The resulting inequality holds true for every $\rho > 0$; minimizing its the right hand side in $\rho > 0$, we arrive at (5.3.69). \square

3⁰. Relation (5.3.69) implies that

$$\mathbf{E} \left\{ \max_{u \in Z} D(\xi^t, u) \right\} \leq 3\Omega^2 + 3\sigma^2 \sum_{\tau=1}^t \gamma_\tau^2,$$

which combines with (5.3.66) to yield (5.3.61). The latter relation, in turn, implies immediately the “in particular” part of Theorem. \square

5.3.7.3 Refinement

Same as its deterministic counterparts, Theorem 5.3.6 admits a refinement as follows:

Theorem 5.3.7 *Assume that (5.3.53) takes place, let $\lambda_1, \lambda_2, \dots$ be deterministic reals satisfying (5.3.28), and let*

$$z^t = \frac{\sum_{\tau=1}^t \lambda_\tau z_\tau}{\sum_{\tau=1}^t \lambda_\tau}.$$

Then

$$\begin{aligned} \mathbf{E}_{[\xi_1; \dots; \xi_t] \sim P \times \dots \times P} \{ \epsilon_{\text{sad}}(z^t) \} &\leq \frac{7}{2} \frac{\frac{\lambda_t}{\gamma_t} \Omega_*^2 + \sigma^2 \sum_{\tau=1}^t \lambda_\tau \gamma_\tau}{\Lambda_t}, \\ \sigma^2 &= \sup_{z \in Z} \mathbf{E}_{\xi \sim P} \{ \|G(z, \xi)\|_*^2 \}, \quad \Lambda_t = \sum_{\tau=1}^t \lambda_\tau. \end{aligned} \tag{5.3.70}$$

In particular, setting

$$\gamma_t = \frac{\Omega_*}{\sigma\sqrt{t}}, \quad t = 1, 2, \dots \tag{5.3.71}$$

and

$$\lambda_t = t^p, \quad t = 1, 2, \dots \tag{5.3.72}$$

where $p > -1/2$, we ensure for every $t = 1, 2, \dots$ the relation

$$\mathbf{E}\{\epsilon_{\text{sad}}(z^t)\} \leq C_p \frac{\Omega_* \sigma}{\sqrt{t}}, \quad (5.3.73)$$

with C_p depending solely and continuously on $p > -1/2$.

Proof. 1⁰. Same as in the proof of Theorem 5.3.5, we have

$$\forall (u \in Z) : \sum_{\tau=1}^t \lambda_\tau \langle G(z_\tau, \xi_\tau), z_\tau - u \rangle \leq \frac{\frac{\lambda_t}{\gamma_t} \Omega_*^2 + \sum_{\tau=1}^t \lambda_\tau \gamma_\tau \|G(z_\tau, \xi_\tau)\|_*^2}{2}.$$

Setting

$$\Delta_\tau = G(z_\tau, \xi_\tau) - F(z_\tau),$$

we can rewrite the above relation as

$$\sum_{\tau=1}^t \lambda_\tau \langle F(z_\tau), z_\tau - u \rangle \leq \frac{\frac{\lambda_t}{\gamma_t} \Omega_*^2 + \sum_{\tau=1}^t \lambda_\tau \gamma_\tau \|G(z_\tau, \xi_\tau)\|_*^2}{2} + \underbrace{\sum_{\tau=1}^t \lambda_\tau \langle \Delta_\tau, u - z_\tau \rangle}_{D(\xi^t, u)}, \quad (5.3.74)$$

where $\xi^t = [\xi_1; \dots; \xi_t]$. It follows that

$$\max_{u \in Z} \sum_{\tau=1}^t \lambda_\tau \langle F(z_\tau), z_\tau - u \rangle \leq \frac{\frac{\lambda_t}{\gamma_t} \Omega_*^2 + \sum_{\tau=1}^t \lambda_\tau \gamma_\tau \|G(z_\tau, \xi_\tau)\|_*^2}{2} + \max_{u \in Z} D(\xi^t, u).$$

As we have seen when proving Theorem 5.3.4, the left hand side in the latter relation is $\geq \Lambda_t \epsilon_{\text{sad}}(z^t)$, so that

$$\Lambda_t \epsilon_{\text{sad}}(z^t) \leq \frac{\frac{\lambda_t}{\gamma_t} \Omega_*^2 + \sum_{\tau=1}^t \lambda_\tau \gamma_\tau \|G(z_\tau, \xi_\tau)\|_*^2}{2} + \max_{u \in Z} D(\xi^t, u). \quad (5.3.75)$$

Same as in the proof of Theorem 5.3.6, the latter relation implies that

$$\Lambda_t \mathbf{E}\{\epsilon_{\text{sad}}(z^t)\} \leq \frac{\frac{\lambda_t}{\gamma_t} \Omega_*^2 + \sigma^2 \sum_{\tau=1}^t \lambda_\tau \gamma_\tau}{2} + \mathbf{E} \left\{ \max_{u \in Z} D(\xi^t, u) \right\}. \quad (5.3.76)$$

2⁰. Applying Lemma 5.3.2 with λ_τ in the role of γ_τ , we get the first inequality in the following chain:

$$\begin{aligned} \mathbf{E}\{\max_{u \in Z} D(\xi^t, u)\} &\leq 6\sigma\Omega\sqrt{\sum_{\tau=1}^t \lambda_\tau^2} \leq 6\sigma\Omega_*\sqrt{\sum_{\tau=1}^t \lambda_\tau^2} \leq 3 \left[\frac{\lambda_t}{\gamma_t} \Omega_*^2 + \frac{\gamma_t}{\lambda_t} \sigma^2 \sum_{\tau=1}^t \lambda_\tau^2 \right] \\ &\leq 3 \left[\frac{\lambda_t}{\gamma_t} \Omega_*^2 + \sigma^2 \sum_{\tau=1}^t \lambda_\tau \gamma_\tau \right], \end{aligned} \quad (5.3.77)$$

where the concluding inequality is due to $\frac{\gamma_t}{\lambda_t} \lambda_\tau^2 \leq \lambda_\tau \gamma_\tau$ for $\tau \leq t$ (recall that (5.3.28) holds true). (5.3.77) combines with (5.3.76) to imply (5.3.70). The latter relation straightforwardly implies the “In particular” part of Theorem. \square

Illustration: Matrix Game and ℓ_1 minimization with uniform fit. “In the nature” there exist “genuine” stochastic minimization/saddle point problems, those where the cost function is given as expectation

$$\phi(x, y) = \int \Phi(x, y, \xi) dP(\xi)$$

with efficiently computable integrant Φ and a distribution P of ξ we can sample from. This is what happens when the actual cost is affected by random factors (prices, demands, etc.). In such a situation, assuming Φ convex-concave and, say, continuously differentiable in x, y for every ξ , the vector $G(x, y, \xi) = [\frac{\partial \Phi(x, y, \xi)}{\partial x}; -\frac{\partial \Phi(x, y, \xi)}{\partial y}]$ under minimal regularity assumptions satisfies (5.3.58) and (5.3.59). Thus, we can mimic a stochastic first order oracle for the cost function by generating a realization $\xi_t \sim P$ and returning the vector $G(x, y, \xi_t)$ (t is the serial number of the call to the oracle, (x, y) is the corresponding query point). Consequently, we can solve the saddle point problem by the SSPMD algorithm. Note that typically in a situation in question it is extremely difficult, if at all possible, to compute ϕ in a closed analytic form and thus to mimic a deterministic first order oracle for ϕ . We may say that in the “genuine” stochastic saddle point problems, the necessity in stochastic oracle-oriented algorithms, like SSPMD, is genuine as well. This being said, we intend to illustrate the potential of stochastic oracles and associated algorithms by an example of a completely different flavour, where the saddle point problem to be solved is a simply-looking deterministic one, and the rationale for randomization is the desire to reduce its computational complexity.

The problem we are interested in is a *bilinear saddle point problem on the direct product of two simplexes*:

$$\min_{x \in \Delta_n} \max_{y \in \Delta_m} \phi(x, y) := a^T x + y^T A x + b^T y.$$

Note that due to the structure of Δ_n and Δ_m , we can reduce the situation to the one where $a = 0, b = 0$; indeed, it suffices to replace the original A with the matrix

$$A + [\underbrace{1; \dots; 1}_m] a^T + b [\underbrace{1; \dots; 1}_n]^T.$$

By this reason, and in order to simplify notation, we pose the problem of interest as

$$\min_{x \in \Delta_n} \max_{y \in \Delta_m} \phi(x, y) := y^T A x. \quad (5.3.78)$$

This problem arises in several applications, e.g., in matrix games and ℓ_1 minimization under ℓ_∞ -loss.

Matrix game interpretation of (5.3.78) (this problem is usually even called “matrix game”) is as follows:

There are two players, every one selecting his action from a given finite set; w.l.o.g., we can assume that the actions of the first player form the set $J = \{1, \dots, n\}$, and the actions of the second player form the set $I = \{1, \dots, m\}$. If the first player chooses an action $j \in J$, and the second – an action $i \in I$, the first pays to the second a_{ij} , where the matrix $A = [a_{ij}]_{\substack{1 \leq i \leq m \\ 1 \leq j \leq n}}$, is known in advance to both the players. What should rational players (interested to reduce their losses and to increase their gains) do?

The answer is easy when the matrix has a saddle point – there exists (i_*, j_*) such that the entry a_{i_*,j_*} is the minimal one in its row and the maximal one in its column. You can immediately verify that in this case the maximal entry in every column is at least a_{i_*,j_*} , and the minimal entry in every row is at most a_{i_*,j_*} , meaning that when the first player chooses some action j , the second one can take from him at least a_{i_*,j_*} , and if the second player chooses an action i , the first can pay to the second at most a_{i_*,j_*} . The bottom line is that rational players should stick to choices j_* and i_* ; in this case, no one of them has an incentive to change his choice, provided that the adversary sticks to his choice. We see that if A has a saddle point, the rational players should use its components as their choices. Now, what to do if, as it typically is the case, A has no saddle point? In this case there is no well-defined solution to the game as it was posed – there is no pair (i_*, j_*) such that the rational players will use, as their respective choices, i_* and j_* . However, there still is a meaningful definition of a solution to a slightly modified game – one in *mixed strategies*, due to von Neumann and Morgenstern. Specifically, assume that the players play the game not just once, but many times, and are interested in their average, over the time, losses. In this case, they can stick to randomized choices, where at t -th tour, the first player draws his choice ξ from J according to some probability distribution on J (this distribution is just a vector $x \in \Delta_n$, called *mixed strategy* of the first player), and the second player draws his choice η according to some probability distribution on I (represented by a vector $y \in \Delta_m$, the mixed strategy of the second player). In this case, assuming the draws independent of each other, the expected per-tour loss of the first player (which is also the expected per-tour gain of the second player) is $y^T A x$. Rational players would now look for a saddle point of $\phi(x, y) = y^T A x$ on the domain $\Delta_n \times \Delta_m$ comprised of pairs of mixed strategies, and in this domain the saddle point does exist.

ℓ_1 minimization with ℓ_∞ fit is the problem $\min_{\xi: \|\xi\|_1 \leq 1} \|B\xi - b\|_\infty$, or, in the saddle point form, $\min_{\|\xi\|_1 \leq 1} \max_{\|\eta\|_1 \leq 1} \eta^T [B\xi - b]$. Representing $\xi = p - q$, $\eta = r - s$ with nonnegative p, q, r, s , the problem can be rewritten in terms of $x = [p; q]$, $y = [r; s]$, specifically, as

$$\min_{x \in \Delta_n} \max_{y \in \Delta_m} [y^T A x - y^T a] \quad [A = \begin{bmatrix} B & -B \\ -B & B \end{bmatrix}, a = [b; -b]]$$

where $n = 2\dim \xi$, $m = 2\dim \eta$. As we remember, the resulting problem can be represented in the form of (5.3.78).

Problem (5.3.78) is a nice bilinear saddle point problem. The best known so far algorithms for solving large-scale instances of this problem within a medium accuracy find an ϵ -solution (x_ϵ, y_ϵ) ,

$$\epsilon_{\text{sad}}(x_\epsilon, y_\epsilon) \leq \epsilon$$

in $O(1)\sqrt{\ln(m)\ln(n)} \frac{\|A\|_\infty + \epsilon}{\epsilon}$ steps, where $\|A\|_\infty$ is the maximal magnitude of entries in A . A single step requires $O(1)$ matrix-vector multiplications involving the matrices A and A^T , plus $O(1)(m + n)$ operations of computational overhead. Thus, the overall arithmetic cost of an ϵ -solution is

$$\mathcal{I}_{\text{det}} = O(1) \frac{\mathcal{A} + \epsilon}{\epsilon} \mathcal{M}$$

arithmetic operations (a.o.), where \mathcal{M} is the maximum of $m+n$ and the arithmetic cost of a pair of matrix-vector multiplications, one involving A and the other one involving A^T . The outlined bound is achieved, e.g., by the Mirror Prox algorithm with Simplex setup, see section 5.5.4.3.

For dense $m \times n$ matrices A with no specific structure, the quantity \mathcal{M} is $O(mn)$, so that the above complexity bound is quadratic in m, n and can become prohibitively large when m, n are really large. Could we do better? The answer is “yes” — we can replace time-consuming in the large scale case precise matrix-vector multiplication by its computationally cheap randomized version.

A stochastic oracle for matrix-vector multiplication can be built as follows. In order to estimate Bx , we can treat the vector $\frac{1}{\|x\|_1} \text{abs}[x]$ as a probability distribution on the set J of indexes of columns in B , draw from this distribution a random index j and return the vector $G = \|x\|_1 \text{sign}(x_j) \text{Col}_j[B]$. We clearly have

$$\mathbf{E}\{G\} = Bx$$

— our oracle indeed is unbiased.

If you think of how could you implement this procedure, you immediately realize that all you need is access to a generator producing a uniformly distributed on $[0, 1]$ random number ξ ; the output of the outlined stochastic oracle will be a deterministic function of x and ξ , as required in our model of stochastic oracle. Moreover, the number of operations to build G modulo generating ξ is dominated by $\dim x$ operations to generate j and the effort to extract from B a column given its number, which we from now on will assume to be $O(1)m$, m being the dimension of the column. Thus, the arithmetic cost of calling the oracle in the case of $m \times n$ matrix B is $O(1)(m + n)$ a.o. (note that for all practical purposes, generating ξ takes just $O(1)$ a.o.). Note also that the output G of our stochastic oracle satisfies

$$\|G\|_\infty \leq \|x\|_1 \|B\|_\infty, \quad \|B\|_\infty = \max_{i,j} |B_{ij}|.$$

Now, the vector field $F(x, y)$ associated with the bilinear saddle point problem (5.3.78) is

$$F(x, y) = [F_x(y) = A^T y; F_y(x) = -Ax] = \mathcal{A}[x; z], \quad \mathcal{A} = \begin{bmatrix} & A^T \\ -A & \end{bmatrix}.$$

We see that computing $F(x, y)$ at a point $(x, y) \in \Delta_n \times \Delta_m$ can be randomized — we can get an unbiased estimate $G(x, y, \xi)$ of $F(x, y)$ at the cost of $O(1)(m + n)$ a.o., and the estimate satisfies the condition

$$\|G(x, y, \xi)\|_\infty \leq \|A\|_\infty. \tag{5.3.79}$$

Indeed, in order to get G given $(x, y) \in \Delta_n \times \Delta_m$, we treat x as a probability distribution on $\{1, \dots, n\}$, draw an index j from this distribution and extract from A the corresponding column $\text{Col}_j[A]$. Similarly, we treat y as a probability distribution on $\{1, \dots, m\}$, draw from this distribution an index i and extract from A^T the corresponding column $\text{Col}_i[A^T]$. The formula for G is just

$$G = [\text{Col}_i[A^T]; -\text{Col}_j[A]].$$

5.3.7.4 Solving (5.3.78) via Stochastic Saddle Point Mirror Descent.

A natural setup for SSPMD as applied to (5.3.78) is obtained by combining the Entropy setups for $X = \Delta_n$ and $Y = \Delta_m$. Specifically, let us define the norm $\|\cdot\|$ on the embedding space \mathbf{R}^{m+n} of $\Delta_n \times \Delta_m$ as

$$\|[x; y]\| = \sqrt{\alpha \|x\|_1^2 + \beta \|y\|_1^2},$$

and the DGF $\omega(x, y)$ for $Z = \Delta_n \times \Delta_m$ as

$$\omega(x, y) = \alpha \sum_{j=1}^n x_j \ln(x_j) + \beta \sum_{i=1}^m y_i \ln(y_i),$$

where α, β are positive weight coefficients. With this choice of the norm and the DGF, we clearly ensure that ω is strongly convex modulus 1 with respect to $\|\cdot\|$, while the ω -radius of $Z = \Delta_n \times \Delta_m$ is given by

$$\Omega^2 = 2[\alpha \ln(n) + \beta \ln(m)],$$

provided that $m, n \geq 3$, which we assume from now on. Finally, the norm conjugate to $\|\cdot\|$ is

$$\|[\xi; \eta]\|_* = \sqrt{\alpha^{-1} \|\xi\|_\infty^2 + \beta^{-1} \|\eta\|_\infty^2}$$

(why?), so that the quantity σ defined in (5.3.61) is, in view of (5.3.79), bounded from above by

$$\hat{\sigma} = \|A\|_\infty \sqrt{\frac{1}{\alpha} + \frac{1}{\beta}}.$$

It makes sense to choose α, β in order to minimize the error bound (5.3.63), or, which is the same, to minimize $\Omega\sigma$. One of the optimizers (they clearly are unique up to proportionality) is

$$\alpha = \frac{1}{\sqrt{\ln(n)} [\sqrt{\ln(n)} + \sqrt{\ln(m)}]}, \quad \beta = \frac{1}{\sqrt{\ln(m)} [\sqrt{\ln(n)} + \sqrt{\ln(m)}]},$$

which results in

$$\Omega = \sqrt{2}, \quad \Omega\sigma = \sqrt{2} \left[\sqrt{\ln(n)} + \sqrt{\ln(m)} \right] \|A\|_\infty.$$

As a result, the efficiency estimate (5.3.63) takes the form

$$\mathbf{E}\{\epsilon_{\text{sad}}(x^T, y^T)\} \leq \frac{6\Omega\sigma}{\sqrt{T}} = 6\sqrt{2} \left[\sqrt{\ln(n)} + \sqrt{\ln(m)} \right] \frac{\|A\|_\infty}{\sqrt{T}}. \quad (5.3.80)$$

Discussion. The number of steps T needed to make the right hand side in the bound (5.3.80) at most a desired $\epsilon \leq \|A\|_\infty$ is

$$N(\epsilon) = O(1) \ln(mn) \left(\frac{\|A\|_\infty}{\epsilon} \right)^2,$$

and the total arithmetic cost of these $N(\epsilon)$ steps is

$$\mathcal{C}_{\text{rand}} = O(1) \ln(mn)(m+n) \left(\frac{\|A\|_\infty}{\epsilon} \right)^2 \text{ a.o.}$$

Recall that for a dense general-type matrix A , to solve (5.3.78) within accuracy ϵ deterministically costs

$$\mathcal{C}_{\text{det}}(\epsilon) = O(1) \sqrt{\ln(m) \ln(n)} mn \left(\frac{\|A\|_\infty}{\epsilon} \right) \text{ a.o.}$$

We see that as far as the dependence of the arithmetic cost of an ϵ -solution on the desired relative accuracy $\delta := \epsilon/\|A\|_\infty$ is concerned, the deterministic algorithm significantly outperforms the randomized one. At the same time, the dependence of the cost on m, n is much better for the randomized algorithm. As a result, when a moderate and fixed relative accuracy is sought and m, n are large, the randomized algorithm by far outperforms the deterministic one.

A remarkable property of the above result is as follows: in order to find a solution of a fixed relative accuracy δ , the randomized algorithm carries out $O(\ln(mn)\delta^{-2})$ steps, and at every step inspects $m+n$ entries of the matrix A (those in a randomly selected row and in a randomly selected column). Thus, the total number of data entries inspected by the algorithm in course of building a solution of relative accuracy δ is $O(\ln(mn)(m+n)\delta^{-2})$, which for large m, n is incomparably less than the total number mn of entries in A . We see that our algorithm demonstrates what is called *sublinear time behaviour* – returns a good approximate solution by inspecting a negligible, for $m = O(n)$ large, and tending to zero as $m = O(n) \rightarrow \infty$, part of problem's data⁷.

Assessing the quality. Strictly speaking, the outlined comparison of the deterministic and the randomized algorithms for (5.3.78) is not completely fair: with the deterministic algorithm, we know for sure that the solution we end up with indeed is an ϵ -solution to (5.3.78), while with the randomized algorithm we know only that the expected inaccuracy of our (random!) approximate solution is $\leq \epsilon$; nobody told us that a realization of this random solution \hat{z} we actually get solves the problem within accuracy about ϵ . If we were able to quantify the actual accuracy of \hat{z} by computing $\epsilon_{\text{sad}}(\hat{z})$, we could circumvent this difficulty as follows: let us choose the number T of steps in our randomized algorithm in such a way that the resulting expected inaccuracy is $\leq \epsilon/2$. After \hat{z} is built, we compute $\epsilon_{\text{sad}}(\hat{z})$; if this quantity is $\leq \epsilon$, we terminate with ϵ -solution to (5.3.78) at hand. Otherwise (this “otherwise” may happen with probability $\leq 1/2$ only, since the inaccuracy is nonnegative, and the expected inaccuracy is $\leq \epsilon/2$) we rerun the algorithm, check the quality of the resulting solution, rerun the algorithm again when this quality still is worse than ϵ , and so on, until either an ϵ -solution is found, or a chosen in advance maximum number N of reruns is reached. With this approach, in order to ensure generating an ϵ -solution to the problem with probability at least $1 - \beta$ we need $N = O(\ln(1/\beta))$; thus, even for pretty small β , N is just a moderate constant⁸, and the randomized algorithm still can outperform by far its 100%-reliable deterministic counterpart. The difficulty, however, is that in order to compute $\epsilon_{\text{sad}}(\hat{z})$, we need to know $F(z)$ exactly, since in our case

$$\epsilon_{\text{sad}}(x, y) = \max_{i \leq m} [Ax]_i - \min_{j \leq n} [A^T y]_j,$$

⁷The first sublinear time algorithm for matrix games was proposed by M. Grigoriadis and L. Khachiyan in 1995 [13]. In retrospect, their algorithm is pretty close, although not identical, to SSPMD; both algorithms share the same complexity bound. In [13] it is also proved that in the situation in question randomization is essential: no deterministic algorithm for solving matrix games can exhibit sublinear time behaviour.

⁸e.g., $N = 20$ for $\beta = 1.\text{e-}6$ and $N = 40$ for $\beta = 1.\text{e-}12$; note that $\beta = 1.\text{e-}12$ is, for all practical purposes, the same as $\beta = 0$.

and even a *single* computation of F could be too expensive (and will definitely destroy the sublinear time behaviour). Can we circumvent somehow the difficulty? The answer is positive. Indeed, note that

A. For a general bilinear saddle point problem

$$\min_{x \in X} \max_{y \in Y} [a^T x + b^T y + y^T A x]$$

the operator F is of a very special structure:

$$F(x, y) = [A^T y + a; -Ax - b] = \mathcal{A}[x; y] + c,$$

where the matrix $\mathcal{A} = \begin{bmatrix} & A^T \\ -A & \end{bmatrix}$ is skew symmetric;

B. The randomized computation of F we have used in the case of (5.3.78) also is of a very specific structure, specifically, as follows: we associate with $z \in Z = X \times Y$ a probability distribution P_z which is supported on Z and is such that if $\zeta \sim P_z$, then $\mathbf{E}\{\zeta\} = z$, and our stochastic oracle works as follows: in order to produce an unbiased estimate of $F(z)$, it draws a realization ζ from P_z and returns $G = F(\zeta)$.

Specifically, in the case of (5.3.78), $z = (x, y) \in \Delta_n \times \Delta_m$, and $P_{x,y}$ is the probability distribution on the mn pairs of vertices (e_j, f_i) of the simplexes Δ_n and Δ_m (e_j are the standard basic orths in \mathbf{R}^n , f_i are the standard basic orths in \mathbf{R}^m) such that the probability of a pair (e_j, f_i) is $x_j y_i$; the expectation of this distribution is exactly z , the expectation of $F(\zeta)$, $\zeta \sim P_{x,y}$ is exactly $F(z)$. The advantage of this particular distribution P_z is that it “sits” on pairs of highly sparse vectors (just one nonzero entry in every member of a pair), which makes it easy to compute F at a realization drawn from this distribution.

The point is that when A and B are met, we can modify the SSPMD algorithm, preserving its efficiency estimate and making $F(z^t)$ readily available for every t . Specifically, at a step τ of the algorithm, we have at our disposal a realization ζ_τ of the distribution P_{z_τ} which we used to compute $G(z_\tau, \xi_\tau) = F(\zeta_\tau)$. Now let us replace the rule for generating approximate solutions, which originally was

$$z^t = \bar{z}^t := \left[\sum_{\tau=1}^t \gamma_\tau \right]^{-1} \sum_{\tau=1}^t \gamma_\tau z_\tau,$$

with the rule

$$z^t = \hat{z}^t := \left[\sum_{\tau=1}^t \gamma_\tau \right]^{-1} \sum_{\tau=1}^t \gamma_\tau \zeta_\tau.$$

since $\zeta_\tau \sim P_{z_\tau}$ and the latter distribution is supported on Z , we still have $z^t \in Z$. Further, we have computed $F(\zeta_\tau)$ exactly, and since F is affine, we just have

$$F(z^t) = \left[\sum_{\tau=1}^t \gamma_\tau \right]^{-1} \sum_{\tau=1}^t \gamma_\tau F(\zeta_\tau),$$

meaning that sequential computing $F(z^t)$ costs us just additional $O(m + n)$ a.o. per step and thus increases the overall complexity by just a close to 1 absolute constant factor. What is

not immediately clear, is why the new rule for generating approximate solutions preserves the efficiency estimate. The explanation is as follows (work it out in full details by yourself): The proof of Theorem 5.3.6 started from the observation that for every $u \in Z$ we have

$$\sum_{\tau=1}^t \gamma_\tau \langle G(z_\tau, \xi_\tau), z_\tau - u \rangle \leq B_t(\xi^t) \quad (!)$$

with some particular $B_t(\xi^t)$; we then verified that the supremum of the left hand side in this inequality in $u \in Z$ is $\geq \left[\sum_{\tau=1}^t \gamma_\tau \right] \epsilon_{\text{sad}}(\bar{z}^t) - C_t(\xi^t)$ with certain explicit $C_t(\cdot)$. As a result, we got the inequality

$$\left[\sum_{\tau=1}^t \gamma_\tau \right] \epsilon_{\text{sad}}(\bar{z}^t) \leq B_t(\xi^t) + C_t(\xi^t), \quad (!!)$$

which, upon taking the expectations, led us to the desired efficiency estimate. Now we should act as follows: it is immediately seen that (!) remains valid and reads

$$\sum_{\tau=1}^t \gamma_\tau \langle F(\zeta_t), z_t - u \rangle \leq B_t(\xi^t),$$

whence

$$\sum_{\tau=1}^t \gamma_\tau \langle F(\zeta_t), \zeta_t - u \rangle \leq B_t(\xi^T) + D_t(\xi^t),$$

where

$$D_t(\xi^t) = \sum_{\tau=1}^t \gamma_\tau \langle F(\zeta_t), \zeta_t - z_t \rangle,$$

which, exactly as in the original proof, leads us to a slightly modified version of (!!), specifically, to

$$\left[\sum_{\tau=1}^t \gamma_\tau \right] \epsilon_{\text{sad}}(\hat{z}^t) \leq B_t(\xi^t) + C_t(\xi^t) + D_t(\xi^t)$$

with the same as in the original proof upper bound on $\mathbf{E}\{B_t(\xi^t) + C_t(\xi^t)\}$. When taking expectations of the both sides of this inequality, we get exactly the same upper bound on $\mathbf{E}\{\epsilon_{\text{sad}}(\dots)\}$ as in the original proof due to the following immediate observation:

$$\mathbf{E}\{D_t(\xi^t)\} = 0.$$

Indeed, by construction, ζ_τ is a deterministic function of ξ^τ , z_τ is a deterministic function of $\xi^{\tau-1}$, and the conditional, $\xi^{\tau-1}$ being fixed, distribution of ζ_τ is P_{z_τ} . We now have

$$\langle F(\zeta_\tau), \zeta_\tau - z_\tau \rangle = \langle \mathcal{A}\zeta_\tau + c, \zeta_\tau - z_\tau \rangle = -\langle \mathcal{A}\zeta_\tau, z_\tau \rangle + \langle c, \zeta_\tau - z_\tau \rangle$$

where we used the fact that \mathcal{A} is skew symmetric. Now, the quantity

$$-\langle \mathcal{A}\zeta_\tau, z_\tau \rangle + \langle c, \zeta_\tau - z_\tau \rangle$$

is a deterministic function of ξ^τ , and among the terms participating in the latter formula the only one which indeed depends on ξ_τ is ζ_τ . Since the dependence of the

quantity on ζ_τ is affine, the conditional, $\xi^{\tau-1}$ being fixed, expectation of it w.r.t. ξ_τ is

$$\langle \mathcal{A} \underbrace{\mathbf{E}_{\xi_\tau}\{\zeta_\tau\}}_{z_\tau}, z_\tau + \langle c, \mathbf{E}_{\xi_\tau}\{\zeta_\tau\} - z_\tau \rangle = -\langle \mathcal{A}z_\tau, z_\tau \rangle + \langle c, z_\tau - z_\tau \rangle = 0$$

(we again used the fact that \mathcal{A} is skew symmetric). Thus, $\mathbf{E}\{\langle F(\zeta_\tau), \zeta_\tau - z_\tau \rangle\} = 0$, whence $\mathbf{E}\{D_t(\xi^t)\} = 0$ as well.

Operational Exercise 5.3.1 The “covering story” for your tasks is as follows:

There are n houses in a city; i -th house contains wealth w_i . A burglar chooses a house, secretly arrives there and starts his business. A policeman also chooses his location alongside one of the houses; the probability for the policeman to catch the burglar is a known function of the distance d between their locations, say, it is $\exp\{-cd\}$ with some coefficient $c > 0$ (you may think, e.g., that when the burglary starts, an alarm sounds or 911 is called, and the chances of catching the burglar depend on how long it takes from the policeman to arrive at the scene). This cat and mouse game takes place every night. The goal is to find numerically near-optimal mixed strategies of the burglar, who is interested to maximize his expected “profit” $[1 - \exp\{-cd(i, j)\}]w_i$ (i is the house chosen by the burglar, j is the house at which the policeman is located), and of the policeman who has a completely opposite interest.

Remark: Note that the sizes of problems of the outlined type can be really huge. For example, let there be 1000 locations, and 3 burglars and 3 policemen instead of one burglar and one policeman. Now there are about $100^3/3!$ “locations” of the burglar side, and about $1000^3/3!$ “locations” of the police side, which makes the sizes of A well above 10^8 .

The model of the situation is a matrix game where the $n \times n$ matrix A has the entries

$$[1 - \exp\{-cd(i, j)\}]w_i,$$

$d(i, j)$ being the distance between the locations i and j , $1 \leq i, j \leq n$. In the matrix game associated with A , the policeman chooses mixed strategy $x \in \Delta_n$, and the burglar chooses a mixed strategy $y \in \Delta_n$. Your task is to get an ϵ -approximate saddle point. We assume for the sake of definiteness that the houses are located at the nodes of a square $k \times k$ grid, and $c = 2/D$, where D is the maximal distance between a pair of houses. The distance $d(i, j)$ is either the Euclidean distance $\|P_i - P_j\|_2$ between the locations P_i, P_j of i -th and j -th houses on the 2D plane, or $d(i, j) = \|P_i - P_j\|_1$ (“Manhattan distance” corresponding to rectangular road network).

Task: Solve the problem within accuracy ϵ around 1.e-3 for $k = 100$ (i.e., $n = k^2 = 10000$). You may play with

- “profile of wealth” w . For example, in my experiment reported on fig. 5.1, I used $w_i = \max[p_i/(k-1), q_i/(k-1)]$, where p_i and q_i are the coordinates of i -th house (in my experiment the houses were located at the 2D points with integer coordinates p, q varying from 0 to $k-1$). You are welcome to use other profiles as well.
- *distance between houses* – Euclidean or Manhattan one (I used the Manhattan distance).

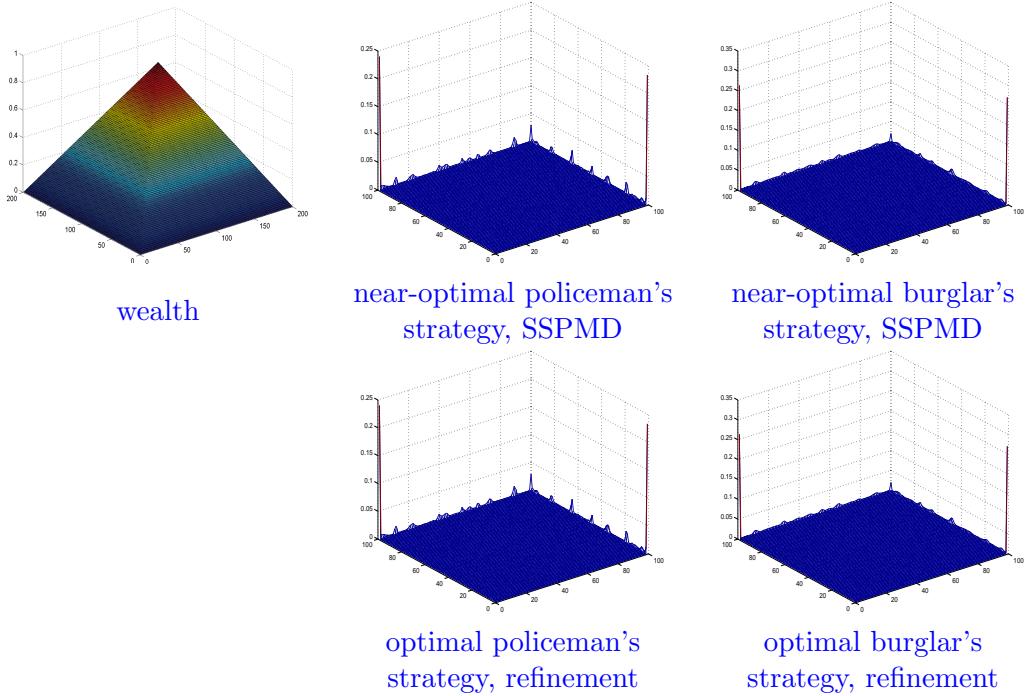


Figure 5.1: My experiment: 100×100 square grid of houses, $T = 50,000$ steps of SSPMD, the stepsize policy (5.3.81) with $\alpha = 20$. At the resulting solution (\bar{x}, \bar{y}) , $\max_{y \in \Delta_n} [y^T A \bar{x}] = 0.5016$, $\min_{x \in \Delta_n} [\bar{y}^T A x] = 0.4985$, meaning that the saddle point value is in-between 0.4985 and 0.5016, while the sum of inaccuracies, in terms of the respective objectives, of the primal solution \bar{x} and the dual solution \bar{y} is at most $\epsilon_{\text{sad}}(\bar{x}, \bar{y}) = 0.5016 - 0.4985 = 0.0031$. It took 1857 sec to compute this solution on my laptop. It took just 3.7 sec more to find a refined solution (\hat{x}, \hat{y}) such that $\epsilon_{\text{sad}}(\hat{x}, \hat{y}) \leq 9.\text{e-}9$, and within 6 decimals after the dot one has $\max_{y \in \Delta_n} [y^T A \hat{x}] = 0.499998$, $\min_{x \in \Delta_n} [\hat{y}^T A x] = 0.499998$. It follows that up to 6 accuracy digits, the saddle point value is 0.499998 (not exactly 0.5!).

- The number of steps T and the stepsize policy. For example, along with the constant stepsize policy (5.3.62), you can try other policies, e.g.,

$$\gamma_t = \alpha \frac{\Omega}{\sigma \sqrt{t}}, \quad 1 \leq t \leq T. \quad (5.3.81)$$

where α is an “acceleration coefficient” which you could tune experimentally (I used $\alpha = 20$, without much tuning). What is the theoretical efficiency estimate for this policy?

- *Looking at your approximate solution, how could you try to improve its quality at a low computational cost? Get an idea, implement it and look how it works.

You could also try to guess in advance what are good mixed strategies of the players and then compare your guesses with the results of your computations.

5.3.8 Mirror Descent and Stochastic Online Regret Minimization

5.3.8.1 Stochastic online regret minimization: problem's formulation

Consider a stochastic analogy of online regret minimization setup considered in section 5.3.5, specifically, as follows. We are given

1. a convex compact set X in Euclidean space E , with (X, E) equipped with proximal setup – a norm $\|\cdot\|$ on E and a compatible with this norm DGF $\omega(\cdot)$, which we assume to be continuously differentiable on X ;
2. a family \mathcal{F} of convex and Lipschitz continuous functions on X ;
3. access to *stochastic oracle* as follows: at t -th call to the oracle, the input to the oracle being a query point $x_t \in X$ and a function $f_t(\cdot) \in \mathcal{F}$, the oracle returns vector $G_t = G_t[f_t(\cdot), x_t, \xi_t]$ and a real $g_t = g_t[f_t(\cdot), x_t, \xi_t]$, where for every t $G_t[f(\cdot), x, \xi]$ and $g_t[f(\cdot), x, \xi]$ are deterministic functions of $f \in \mathcal{F}$, $x \in X$ and a realization ξ of “oracle's noise.” It is further assumed that
 - (a) $\{\xi_t \sim P : t \geq 0\}$ is an i.i.d. sequence of random “oracle noises” (noise distribution P is not known in advance);
 - (b) $f_t(\cdot) \in \mathcal{F}$ is selected “by nature” and depends solely on t and the collection $\xi^{t-1} = (\xi_0, \xi_1, \dots, \xi_{t-1})$ of oracle noises at steps preceding the step t :

$$f_t(\cdot) = F_t(\xi^{t-1}, \cdot),$$

where $F_t(\xi^{t-1}, x)$ is a *legitimate* deterministic function of ξ^{t-1} and $x \in X$, with “legitimate” meaning that $F_t(\xi^{t-1}, x)$ as a function of x belongs to \mathcal{F} for all ξ^{t-1} . Sequence $\{F_t(\cdot, \cdot) : t \geq 1\}$ is selected by nature prior to our actions. In the sequel, we refer to such a sequence as to *nature's policy*.

Finally, we assume that when the stochastic oracle is called at step t , these are we (the decision maker) who inputs to the stochastic oracle the query point x_t , and this is the nature inputting $f_t(\cdot) = F_t(\xi^{t-1}, \cdot)$; we do not see $f_t(\cdot)$ directly, all we see is the returned by oracle stochastic information $g_t[f_t(\cdot), x_t, \xi_t], G_t[f_t(\cdot), x_t, \xi_t]$ on f_t .

- (c) our actions x_1, x_2, \dots are yielded by a *non-anticipative decision making policy* as follows. At step t we have at our disposal the past search points x_τ , $\tau < t$, along with the answers g_τ, G_τ of the oracle queried at the inputs $(x_\tau, F_\tau(\xi^{\tau-1}, \cdot))$, $1 \leq \tau < t$. x_t should be a deterministic function of t and of the tuple $\{x_\tau, g_\tau, G_\tau : \tau < t\}$; a non-anticipative decision making policy is nothing but the collection of these deterministic functions.

Note that with a non-anticipative decision making policy fixed, the query points x_t become deterministic functions of ξ^{t-1} , $t = 1, 2, \dots$

Our crucial assumption (which is in force everywhere in this section) is that whatever be t , x and $f(\cdot) \in \mathcal{F}$, we have

$$\begin{aligned} (a) \quad \mathbf{E}_{\xi \sim P}\{g_t[f(\cdot), x, \xi]\} &= f(x) \\ (b.1) \quad \mathbf{E}_{\xi \sim P}\{G_t[f(\cdot), x, \xi]\} &\in \partial f(x_t) \\ (b.2) \quad \mathbf{E}_{\xi \sim P}\{\|G_t[f(\cdot), x, \xi]\|_*^2\} &\leq \sigma^2 \end{aligned} \tag{5.3.82}$$

with some known in advance positive constant σ .

Finally, we assume that the toll we are paying in round t is exactly the quantity $g_t[f_t(\cdot), x_t, \xi_t]$, and that our goal is to devise a non-anticipative decision making policy which makes small, as N grows, the expected regret

$$\begin{aligned} & \mathbf{E}_{\xi^N} \left\{ \frac{1}{N} \sum_{t=1}^N g_t[F_t(\xi^{t-1}, \cdot), x_t, \xi_t] \right\} - \min_{x \in X} \mathbf{E}_{\xi^N} \left\{ \frac{1}{N} \sum_{t=1}^N g_t[F_t(\xi^{t-1}, \cdot), x, \xi_t] \right\} \\ = & \mathbf{E}_{\xi^N} \left\{ \frac{1}{N} \sum_{t=1}^N F_t(\xi^{t-1}, x_t) \right\} - \min_{x \in X} \mathbf{E}_{\xi^N} \left\{ \frac{1}{N} \sum_{t=1}^N F_t(\xi^{t-1}, x_t) \right\} \end{aligned} \quad (5.3.83)$$

(the equality here stems from (5.3.82.a) combined with the facts that x_t is a deterministic function of ξ^{t-1} and that $\xi_0, \xi_1, \xi_2, \dots$ are i.i.d.). Note that we want to achieve the outlined goal whatever be nature's policy, that is, whatever be a sequence $\{F_t(\xi^{t-1}, x) : t \geq 1\}$ of legitimate deterministic functions.

5.3.8.2 Minimizing stochastic regret by MD

We have specified the problem of stochastic regret minimization; what is on our agenda now, is to understand how this problem can be solved by Mirror Descent. We intend to use exactly the same algorithm as in section 5.3.5:

1. *Initialization:* Select stepsizes $\{\gamma_t > 0\}_{t \geq 1}$ which are nonincreasing in t and select $x_1 \in X$.
2. *Step $t = 1, 2, \dots$* Given x_t , call the oracle, x_t being the query point, pay the corresponding toll g_t , observe G_t , set

$$x_{t+1} = \text{Prox}_{x_t}(\gamma_t G_t) := \underset{x \in X}{\operatorname{argmin}} [\langle \gamma_t G_t - \omega'(x_t), x \rangle + \omega(x)]$$

and pass to step $t + 1$.

Note that with the selected by the nature sequence $\{F_t(\xi^{t-1}, x) : t \geq 1\}$ of legitimate deterministic functions fixed, x_t are deterministic functions of ξ^{t-1} , $t = 1, 2, \dots$. Note also that the decision making policy yielded by the algorithm indeed is non-anticipative (and even does not require knowledge of tolls g_t).

Convergence analysis. We are about to prove the following straightforward modification of Theorem 5.3.3:

Theorem 5.3.8 *In the notation and under assumptions of this section, whatever be nature's policy, that is, whatever be a sequence $\{F_t(\cdot, \cdot) : t \geq 1\}$ of legitimate deterministic functions, for the induced by this policy trajectory $\{x_t = x_t(\xi^{t-1}) : t \geq 1\}$ of the above algorithm, and for every deterministic sequence $\{\lambda_t > 0 : t \geq 1\}$ such that λ_t/γ_t is nondecreasing in t , it holds for $N = 1, 2, \dots$:*

$$\begin{aligned} & \mathbf{E}_{\xi^N} \left\{ \frac{1}{\Lambda_N} \sum_{t=1}^N g_t[F_t(\xi^{t-1}, \cdot), x_t, \xi_t] \right\} - \min_{x \in X} \mathbf{E}_{\xi^N} \left\{ \frac{1}{\Lambda_N} \sum_{t=1}^N g_t[F_t(\xi^{t-1}, \cdot), x, \xi_t] \right\} \\ \leq & \frac{\frac{\lambda_N}{\gamma_N} \Omega_*^2 + \sigma^2 \sum_{t=1}^N \lambda_t \gamma_t}{2\Lambda_N} \end{aligned} \quad (5.3.84)$$

Here, same as in Theorem 5.3.3,

$$\Lambda_N = \sum_{t=1}^N \lambda_t, \quad \Omega_* = \sup_{x, y \in X} \sqrt{2V_x(y)} \equiv \sup_{x, y \in X} \sqrt{2[\omega(y) - \omega(x) - \langle \omega'(x), y - x \rangle]}.$$

Proof. Repeating word by word the reasoning in the proof of Theorem 5.3.3, we arrive at the relation

$$\sum_{t=1}^N \lambda_t \langle G_t[F_t(\xi^{t-1}, \cdot), x_t, \xi_t], x_t - x \rangle \leq \frac{1}{2} \left[\frac{\lambda_N}{\gamma_N} \Omega_*^2 + \sum_{t=1}^N \lambda_t \gamma_t \|G_t[F_t(\xi^{t-1}, \cdot), x_t, \xi_t]\|_*^2 \right].$$

which holds true for every $x \in X$. Taking expectations of both sides and invoking the fact that $\{\xi_t\}$ are i.i.d. in combination with (5.3.82.b) and the fact that x_t is a deterministic function of ξ^{t-1} , we arrive at

$$\mathbf{E}_{\xi^N} \left\{ \sum_{t=1}^N \lambda_t \langle F'_t(\xi^{t-1}, x_t), x_t - x \rangle \right\} \leq \frac{1}{2} \left[\frac{\lambda_N}{\gamma_N} \Omega_*^2 + \sigma^2 \sum_{t=1}^N \lambda_t \gamma_t \right], \quad (5.3.85)$$

where $F'_t(\xi^{t-1}, x_t)$ is a subgradient, taken w.r.t. x , of the function $F_t(\xi^{t-1}, x)$ at $x = x_t$. Since $F_t(\xi^{t-1}, x)$ is convex in x , we have $F_t(\xi^{t-1}, x_t) - F_t(\xi^{t-1}, x) \leq \langle F'_t(\xi^{t-1}, x_t), x_t - x \rangle$, and therefore (5.3.85) implies that for all $x \in X$ it holds

$$\mathbf{E}_{\xi^N} \left\{ \sum_{t=1}^N \lambda_t [F_t(\xi^{t-1}, x_t) - F_t(\xi^{t-1}, x)] \right\} \leq \frac{1}{2} \left[\frac{\lambda_N}{\gamma_N} \Omega_*^2 + \sigma^2 \sum_{t=1}^N \lambda_t \gamma_t \right],$$

whence also

$$\mathbf{E}_{\xi^N} \left\{ \frac{1}{\Lambda_N} \sum_{t=1}^N \lambda_t F_t(\xi^{t-1}, x_t) \right\} - \min_{x \in X} \mathbf{E}_{\xi^N} \left\{ \frac{1}{\Lambda_N} \sum_{t=1}^N \lambda_t F_t(\xi^{t-1}, x) \right\} \leq \frac{\frac{\lambda_N}{\gamma_N} \Omega_*^2 + \sigma^2 \sum_{t=1}^N \lambda_t \gamma_t}{2\Lambda_N}.$$

Invoking the second representation of the expected regret in (5.3.83), the resulting relation is exactly (5.3.84). \square

Consequences. Let us choose

$$\gamma_t = \frac{\Omega_* \sqrt{1+\alpha}}{\sigma \sqrt{t}}, \quad \lambda_t = t^\alpha,$$

with $\alpha \geq 0$. This choice meets the requirements from the description of our algorithm and from the premise of Theorem 5.3.8, and, similarly to what happened in section 5.3.5, results in

$$\begin{aligned} & \forall (N \geq \alpha) : \\ & \mathbf{E}_{\xi^N} \left\{ \frac{1}{\Lambda_N} \sum_{t=1}^N \lambda_t F_t(\xi^{t-1}, x_t) \right\} - \min_{x \in X} \mathbf{E}_{\xi^N} \left\{ \frac{1}{\Lambda_N} \sum_{t=1}^N \lambda_t F_t(\xi^{t-1}, x) \right\} \leq O(1) \frac{\sqrt{1+\alpha} \Omega_* \sigma}{\sqrt{N}}, \end{aligned} \quad (5.3.86)$$

with interpretation completely similar to the interpretation of the bound (5.3.44), see section 5.3.5.

5.3.8.3 Illustration: predicting sequences

To illustrate the stochastic regret minimization, consider *Predicting Sequence* problem as follows:

“The nature” selects a sequence $\{z_t : t \geq 1\}$ with terms z_t from the d -element set $\mathcal{Z}_d = \{1, \dots, d\}$. We observe these terms one by one, and at every time t want to predict z_t via the past observations z_1, \dots, z_{t-1} . Our loss at step t , our prediction being \hat{z}_t , is $\phi(\hat{z}_t, z_t)$, where ϕ is a given *loss function* such that $\phi(z, z) = 0$, $z = 1, \dots, d$, and $\phi(z', z) > 0$ when $z' \neq z$. What we are interested in is to make the average, over time horizon $t = 1, \dots, N$, loss small.

What was just said is a “high level” imprecise setting. Missing details are as follows. We assume that

1. the nature generates sequences z_1, z_2, \dots in a randomized fashion. Specifically, there exists an i.i.d. sequence of “primitive” random variables ζ_t , $t = 0, 1, 2, \dots$ observed one by one by the nature, and z_t is a deterministic function $Z_t(\zeta^{t-1})$ of ζ^{t-1} ⁹. We refer to the collection $Z^\infty = \{Z_t(\cdot) : t \geq 1\}$ as to the *generation policy*.
2. we are allowed for *randomized prediction*, where the predicted value \hat{z}_t of z_t is generated at random according to probability distribution x_t – a vector from the probabilistic simplex $X = \{x \in \mathbf{R}^d : x \geq 0, \sum_{z=1}^d x^z = 1\}$; an entry x_t^z in x_t is the probability for our randomized prediction to take value z , $z = 1, \dots, d$.

We impose on x_t the restriction to be a deterministic function of ζ^{t-1} : $x_t = X_t(\zeta^{t-1})$, and refer to a collection $X^\infty = \{X_t(\cdot) : t \geq 1\}$ of deterministic functions $X_t(\zeta^{t-1})$ taking values in X as to *prediction policy*.

Introducing auxiliary sequence η_0, η_1, \dots of independent of each other and of ζ_t ’s random variables uniformly distributed on $[0, 1)$, we can make our randomized predictions $\hat{z}_t \sim x_t$ deterministic functions of x_t and η_t :

$$\hat{z}_t = \check{Z}(x_t, \eta_t),$$

where, for a probabilistic vector x and a real $\eta \in [0, 1)$, $\check{Z}(x, \eta)$ is the integer $z \in \mathcal{Z}_d$ uniquely defined by the relation $\sum_{\ell=1}^{z-1} x^\ell \leq \eta < \sum_{\ell=1}^z x^\ell$ (think of how would you sample from a probability distribution x on $\{1, \dots, d\}$). As a result, setting $\xi_t = [\zeta_t; \eta_t]$ and thinking about the i.i.d. sequence ξ_0, ξ_1, \dots as of the input “driving” generation and prediction, the respective policies being Z^∞ and X^∞ , we see that

- z_t ’s are deterministic functions of $\xi^{t-1} = [\zeta^{t-1}; \eta^{t-1}]$, specifically, $z_t = Z_t(\zeta^{t-1})$;
- x_t ’s are deterministic functions of ζ^{t-1} , specifically, $X_t(\zeta^{t-1})$, and thus are deterministic functions of ξ^{t-1} : $x_t = \tilde{X}_t(\zeta^{t-1})$;
- predictions \hat{z}_t ’s are deterministic functions of (ζ^{t-1}, η_t) and thus – of (ξ^{t-1}, η_t) : $\hat{z}_t = \widehat{Z}_t(\xi^{t-1}, \eta_t)$.

Note that functions Z_t are fully determined by generation policy, while functions \tilde{X}_t and \widehat{Z}_t stem from interaction between the policies for generation and for prediction and are fully determined by these two policies.

Given a generation policy $Z^\infty = \{Z_t(\cdot) : t \geq 1\}$, a prediction policy $X^\infty = \{X_t(\cdot) : t \geq 1\}$ and a time horizon N , we

⁹as always, given a sequence u_0, u_1, u_2, \dots , we denote by $u^t = (u_0, u_1, \dots, u_t)$ the initial fragment, of length $t+1$, of the sequence. Similarly, for a sequence u_1, u_2, \dots , u^t stands for the initial fragment (u_1, \dots, u_t) of length t of the sequence.

- quantify the performance of X^∞ by the quantity

$$\mathbf{E}_{\zeta^N, \eta^N} \left\{ \frac{1}{N} \sum_{t=1}^N \phi(\widehat{Z}_t(\zeta^{t-1}, \eta_t), Z_t(\zeta^{t-1})) \right\},$$

that is, by the averaged over the time horizon expected prediction error as quantified by our loss function ϕ , and

- define the *expected N-step regret* associated with Z^∞ , X^∞ as

$$\begin{aligned} \mathcal{R}_N[X^\infty | Z^\infty] &= \mathbf{E}_{\zeta^N, \eta^N} \left\{ \frac{1}{N} \sum_{t=1}^N \phi(\widehat{Z}_t(\zeta^{t-1}, \eta_t), Z_t(\zeta^{t-1})) \right\} \\ &\quad - \inf_{x \in X} \mathbf{E}_{\zeta^N, \eta^N} \left\{ \frac{1}{N} \sum_{t=1}^N \phi(\widehat{Z}_{t,x}(\zeta^{t-1}, \eta_t), Z_t(\zeta^{t-1})) \right\}, \end{aligned} \quad (5.3.87)$$

where the only yet undefined quantity $\widehat{Z}_{t,x}(\zeta^{t-1}, \eta_t)$ is the “stationary counterpart” of $\widehat{Z}_t(\zeta^{t-1}, \eta_t)$, that is, $\widehat{Z}_{t,x}(\zeta^{t-1}, \eta_t)$ is t -th prediction yielded by interaction between the generation policy Z^∞ and the *stationary* prediction policy $X_x^\infty = \{x_t \equiv x : t \geq 1\}$. Thus, the regret is the excess of the average (over time and “driving input” ξ_0, ξ_1, \dots) prediction error yielded by prediction policy X^∞ , generation policy being Z^∞ , over the similar average error of the best possible under the circumstances (under generation policy Z^∞) stationary ($x_t = x, t \geq 1$) prediction policy.

Put broadly and vaguely, *our goal is to build a prediction policy which, whatever be a generation policy, makes the regret small provided N is large*.

To achieve our goal, we are about to “translate” our prediction problem to the language of stochastic regret minimization as presented in section 5.3.8.1. A sketch of the translation is as follows. Let \mathcal{F} be the set of d linear functions

$$f^\ell(x) = \sum_{z=1}^d \phi(z, \ell) x^z : X \rightarrow \mathbf{R}, \quad \ell = 1, \dots, d;$$

$f^\ell(x)$ is just the expected error of the randomized prediction, the distribution of the prediction being $x \in X$, in the case when the true value to be predicted is ℓ . Note that $f^\ell(\cdot)$ is fully determined by ℓ and “remembers” ℓ (since by our assumption on the loss function we have $f^\ell(\ell) = 0$ and $f^\ell(z) > 0$ when $z \neq \ell$). Consequently, we lose nothing when assuming that the nature, instead of generating a sequence $\{z_t : t \geq 1\}$ of points from \mathcal{Z}_d , generates a sequence of functions $f_t(\cdot) = f^{z_t}(\cdot)$ from \mathcal{F} , $t = 1, 2, \dots$; with this interpretation, a generation policy Z^∞ becomes what was in section 5.3.8.1 called nature’s policy, a policy which specifies t -th selection $f_t(\cdot)$ of nature as deterministically depending on ξ^{t-1} as on a parameter convex (in fact, linear) function $F_t(\xi^{t-1}, \cdot) : X \rightarrow \mathbf{R}$.

Now, at a step t , after the nature shows us z_t , we get full information on the function $f_t(\cdot)$ selected by the nature at this step, and thus know the gradient of this affine function; this gradient can be taken as what in the decision making framework of section 5.3.8.1 was called G_t . Besides this, after the nature reveals z_t , we know which prediction error we have made, and this error can be taken as what in section 5.3.8.1 was called the toll g_t . On a straightforward inspection which we leave to the reader, the outlined translation converts the regret minimization form of our prediction problem into the problem of stochastic online regret minimization considered in section 5.3.8.1 and ensures validity of all assumptions of Theorem 5.3.8. As a result, we conclude that

(!) The MD algorithm from section 5.3.8.2 can be straightforwardly applied to the Sequence Prediction problem; with proper proximal setup (namely, the Simplex one) for the probabilistic simplex X and properly selected stepsize policy, this algorithm yields prediction policy X_{MD}^{∞} which guarantees the regret bound

$$\mathcal{R}_N[X_{\text{MD}}^{\infty} | Z^{\infty}] \leq O(1) \frac{\sqrt{\ln(d+1)} \max_{z,\ell} |\phi(z, \ell)|}{\sqrt{N}}. \quad (5.3.88)$$

whatever be $N = 1, 2, \dots$ and a sequence generation policy Z^{∞} .

We see that MD allows to push regret to 0 as $N \rightarrow \infty$ at an $O(1/\sqrt{N})$ rate and uniformly in sequence generation policies of the nature. As a result, when sequence generating policy used by the nature is “well suited” for stationary prediction, i.e., nature’s policy makes small the comparator

$$\inf_{x \in X} \mathbf{E}_{\zeta^N, \eta^N} \left\{ \frac{1}{N} \sum_{t=1}^N \phi(\hat{Z}_{t,x}(\zeta^{t-1}, \eta_t), Z_t(\zeta^{t-1})) \right\} \quad (5.3.89)$$

in (5.3.87), (!) says that the MD-based prediction policy X_{MD}^{∞} is good provided N is large. In the examples to follow, we use the simplest possible loss function: $\phi(z, \ell) = 0$ when $z = \ell$ and $\phi(z, \ell) = 1$ otherwise.

Example 1: i.i.d. generation policy. Assume that the nature generates z_t by drawing it, independently across t , from some distribution p (let us call this generation policy an i.i.d. one). It is immediately seen that due to the i.i.d. nature of z_1, z_2, \dots , there is no way to predict z_t via z_1, \dots, z_{t-1} too well, even when we know p in advance; specifically, the smallest possible expected prediction error is

$$\epsilon(p) := 1 - \max_{1 \leq z \leq d} p^z,$$

and the corresponding optimal prediction policy is to say all the time that the predicted value of z_t is the most probable value z_* (the index of the largest entry in p). Were p known to us in advance, we could use this trivial optimal predictor from the very beginning. On the other hand, the optimal predictor is stationary, meaning that with the generating policy in question, the comparator is equal to $\epsilon(p)$. As a result, (!) implies that our MD-based predictor policy is “asymptotically optimal” in the i.i.d. case: whatever be an i.i.d. generation policy used by the nature, the average, over time horizon N , expected prediction error of the MD-based prediction can be worse than the best one can get under the circumstances by at most $O(1)\sqrt{\ln(d+1)}/\sqrt{N}$. A good news is that when designing the MD-based policy, we did not assume that the generation policy is an i.i.d. one; in fact, we made no assumptions on this policy at all!

Example 2: predicting an individual sequence. Note that a whatever individual sequence $z^{\infty} = \{z_t : t \geq 1\}$ is given by appropriate generation policy, specifically, a policy where $Z_t(\zeta^{t-1}) \equiv z_t$ is independent of ζ^{t-1} . For this policy and with our simple loss function, the comparator (5.3.89), as is immediately seen, is

$$\epsilon_N[z^{\infty}] = 1 - \max_{1 \leq z \leq d} \frac{\text{Card}(\{t \leq N : z_t = z\})}{N},$$

that is, it is $\epsilon[p^N]$, where p^N is the empirical distribution of z_1, \dots, z_N . We see that out MD-based predictor behaves reasonably well on individual sequences which are “nearly stationary,”

yielding results like: “if the fraction of entries different from 1 in any initial fragment, of length at least N_0 , of z^∞ is at most 0.05, then the percent of errors in the MD-based prediction of the first $N \geq N_0$ entries of z^∞ will be at most $0.05 + O(1)\sqrt{\ln(d+1)/N}$.” Not that much of a claim, but better than nothing!

Of course, there exist “easy to predict” individual sequences (like the alternating sequence 1, 2, 1, 2, 1, 2, ...) where the performance of any stationary predictor is poor; in these cases, we cannot expect much from our MD-based predictor.

5.4 Bundle Mirror algorithm

In spite of being optimal in terms of Information-Based Complexity Theory in the large-scale case, the Mirror Descent algorithm has really poor theoretical rate of convergence, and its typical practical behaviour is as follows: the objective rapidly improves at few first iterations and then the method “gets stuck,” perhaps still far from the optimal value; to get further improvement, one should wait hours, if not days. One way to improve, to some extent, the practical behaviour of the method is to better utilize the first order information accumulated so far. In this respect, the MD itself is poorly organized (cf. comments on SD in the beginning of section 5.2.2): all information accumulated so far is “summarized” in a single entity – the current iterate; we can say that the method is “memoryless.” Computational experience shows that the first order methods for nonsmooth convex optimization which do “have memory,” primarily, the so called *bundle algorithms*, typically perform much better than the memoryless methods like MD. In what follows, we intend to present a bundle version of MD – the *Bundle Mirror* (BM) and the *Truncated Bundle Mirror* (TBM) algorithms for solving the problem

$$f_* = \min_{x \in X} f(x) \quad (\text{CP})$$

Note that BM is in the same relation to Bundle Level method from section 5.2.2 as MD is to SD.

The assumptions on the problem and the setups for these algorithms are exactly the same as for MD; in particular, X is assumed to be bounded (see Standing Assumption, section 5.3.4.2).

5.4.1 Bundle Mirror: Description

The setup for the BM algorithm is exactly the same as for the MD one. As applied to problem (CP), the BM algorithm works as follows.

A. The algorithm generates a sequence of search points, all belonging to X , where the First Order oracle is called, and at every step builds the following entities:

1. the best found so far value of f along with the best found so far search point, which is treated as the current approximate solution built by the method;
2. a (valid) lower bound on the optimal value f_* of the problem.

B. The execution is split in subsequent phases. Phase s , $s = 1, 2, \dots$, is associated with prox-center $c_s \in X^o$ and level $\ell_s \in \mathbf{R}$ such that

- when starting the phase, we already know what is $f(c_s)$, $f'(c_s)$;

- $\ell_s = f_s + \lambda(f^s - f_s)$, where
 - f^s is the best value of the objective known at the time when the phase starts;
 - f_s is the lower bound on f_* we have at our disposal when the phase starts;
 - $\lambda \in (0, 1)$ is a parameter of the method.

In principle, the prox-center c_1 corresponding to the very first phase can be chosen in X^o in an arbitrary fashion; we shall discuss details in the sequel. We start the entire process with computing f , f' at this prox-center, which results in

$$f^1 = f(c_1)$$

and set

$$f_1 = \min_{x \in X} [f(c_1) + (x - c_1)^T f'(x_1)],$$

thus getting the initial lower bound on f_* .

C. Now let us describe a particular phase s . Let

$$\omega_s(x) = \omega(x) - \langle \omega'(c_s), x - c_s \rangle;$$

note that (5.3.3) implies that

$$\begin{aligned} \omega_s(y) &\geq \omega_s(x) + \langle \omega'_s(x), y - x \rangle + \frac{1}{2} \|y - x\|^2 \quad \forall x \in X^o, y \in X. \\ [\omega'_s(x)] &:= \omega'(x) - \omega'(c_s) \end{aligned} \tag{5.4.1}$$

and that c_s is the minimizer of $\omega_s(\cdot)$ on X .

The search points $x_t = x_{t,s}$ of the phase s , $t = 1, 2, \dots$ belong to X^o and are generated according to the following rules:

1. When generating x_t , we already have at our disposal x_{t-1} and a convex compact set $X_{t-1} \subset X$ such that

(o_{t-1}): $X_{t-1} \subset X$ is of the same affine dimension as X , meaning that X_{t-1} contains a neighbourhood (in X) of a point from the relative interior of X ¹⁰

and

$$\begin{aligned} (a_{t-1}) \quad &x \in X \setminus X_{t-1} \Rightarrow f(x) > \ell_s; \\ (b_{t-1}) \quad &x_{t-1} \in \operatorname{argmin}_{X_{t-1}} \omega_s. \end{aligned} \tag{5.4.2}$$

Here $x_0 = c_s$ and, say, $X_0 = X$, which ensures (o_0) and (5.4.2.a₀-b₀).

2. To update (x_{t-1}, X_{t-1}) into (x_t, X_t) , we solve the auxiliary problem

$$\min_x \left\{ g_{t-1}(x) \equiv f(x_{t-1}) + (x - x_{t-1})^T f'(x_{t-1}) : x \in X_{t-1} \right\}. \tag{L_{t-1}}$$

Our subsequent actions depend on the results of this optimization:

¹⁰The reader will lose nothing by assuming that X is full-dimensional – possesses a nonempty interior. Then (o_{t-1}) simply means that X_{t-1} possesses a nonempty interior.

- (a) Let $\tilde{f} \leq +\infty$ be the optimal value in (L_{t-1}) (as always, $\tilde{f} = \infty$ means that the problem is infeasible). Observe that the quantity

$$\hat{f} = \min[\tilde{f}, \ell_s]$$

is a lower bound on f_* : in $X \setminus X_{t-1}$ we have $f(x) > \ell_s$ by (5.4.2.a_{t-1}), while on X_{t-1} we have $f(x) \geq \hat{f}$ due to the inequality $f(x) \geq g_{t-1}(x)$ given by the convexity of f . Thus, $f(x) \geq \min[\ell_s, \hat{f}]$ everywhere on X .

In the case of

$$\hat{f} \geq \ell_s - \theta(\ell_s - f_s) \quad (5.4.3)$$

(“significant” progress in the lower bound), we terminate the phase and update the lower bound on f_* as

$$f_s \mapsto f_{s+1} = \hat{f}.$$

The prox-center c_{s+1} for the new phase can, in principle, be chosen in X^o in an arbitrary fashion (see below).

- (b) When no significant progress in the lower bound is observed, we solve the optimization problem

$$\min_x \{\omega_s(x) : x \in X_{t-1}, g_{t-1}(x) \leq \ell_s\}. \quad (P_{t-1}).$$

This problem is feasible, and moreover, its feasible set is of the same dimension as X .

Indeed, since X_{t-1} is of the same dimension as X , the only case when the set $\{x \in X_{t-1} : g_{t-1}(x) \leq \ell_s\}$ would lack the same property could be the case of $\hat{f} = \min_{x \in X_{t-1}} g_{t-1}(x) \geq \ell_s$, whence $\hat{f} = \ell_s$, and therefore (5.4.3) takes place, which is not the case.

When solving (P_{t-1}) , we get the optimal solution x_t of this problem and compute $f(x_t)$, $f'(x_t)$. It can be proved (see below) that $x_t \in X^o$, as required from all our search points, and that with $\omega'_s(x) := \omega'(x) - \omega'(c_s)$ one has

$$\langle \omega'_s(x_t), x - x_t \rangle \geq 0, \quad \forall (x \in X_{t-1} : g_t(x) \leq \ell_s). \quad (5.4.4)$$

It is possible that we get a “significant” progress in the objective, specifically, that

$$f(x_t) - \ell_s \leq \theta(f^s - \ell_s), \quad (5.4.5)$$

where $\theta \in (0, 1)$ is a parameter of the method. In this case, we again terminate the phase and set

$$f^{s+1} = f(x_t);$$

the lower bound f_{s+1} on the optimal value is chosen as the maximum of f_s and of all lower bounds \hat{f} generated during the phase. The prox-center c_{s+1} for the new phase, same as above, can in principle be chosen in X^o in an arbitrary fashion.

- (c) When (P_{t-1}) is feasible and (5.4.5) does not take place, we continue the phase s , choosing as X_t an arbitrary convex compact set such that

$$\underline{X}_t \equiv \{x \in X_{t-1} : g_{t-1}(x) \leq \ell_s\} \subset X_t \subset \overline{X}_t \equiv \{x \in X : (x - x_t)^T \omega'_s(x_t) \geq 0\}. \quad (5.4.6)$$

Note that we are in the case when (P_{t-1}) is feasible and x_t is the optimal solution to the problem; as we have claimed, this implies (5.4.4), so that

$$\emptyset \neq \underline{X}_t \subset \overline{X}_t,$$

and thus (5.4.6) indeed allows to choose X_t . Since, as we have seen, \underline{X}_t is of the same dimension as X , this property will be automatically inherited by X_t , as required by (o_t). Besides this, every choice of X_t compatible with (5.4.6) ensures (5.4.2.a_t) and (5.4.2.b_t); the first relation is clearly ensured by the left inclusion in (5.4.6) combined with (5.4.2.a_{t-1}) and the fact that $f(x) \geq g_{t-1}(x)$, while the second relation (5.4.2.b_t) follows from the right inclusion in (5.4.6) due to the convexity of $\omega_s(\cdot)$.

Clearing debts. We have claimed that when x_t is well defined, $\omega'_s(x_t)$ exists and (5.4.4) is satisfied. This is evident when $\omega(\cdot)$ is continuously differentiable on the entire X , and needs certain justification otherwise. Here is the justification. W.l.o.g. we can assume that X is full-dimensional, whence, as we have already explained, the set $Y = \{x \in X_{t-1} : g_{t-1}(x) \leq \ell_s\}$ also is full-dimensional (recall that we are in the case when the phase s does not terminate at step t due to significant progress in the lower bound on f_*). Let $\bar{x} \in \text{int } Y$, let $V \subset Y$ be a ball of positive radius ρ centered at \bar{x} , and let $x(r) = x_t + r(\bar{x} - x_t)$, so that the function $\psi(r) = \omega_s(x(r))$ (which clearly is convex) attains its minimum on $[0, 1]$ at $r = 0$. For $0 < r \leq 1$ we have $x(r) \in \text{int } X$, so that $g(r) := \omega'_s(x(r))$ exists. Since $\psi(r) \geq \psi(0)$ and ψ is convex, we have $0 \leq \psi'(r) = \langle g(r), \bar{x} - x_t \rangle$ and besides this, for $\|h\|_2 \leq \rho$ we have $\langle g(r), \bar{x} + h - x(r) \rangle \leq \omega_s(\bar{x} + h) - \omega_s(x(r)) \leq C$, where C is independent of r and h . In other words,

$$\forall (h : \|h\|_2 \leq \rho) : \langle g(r), h \rangle = \langle g(r), \bar{x} + h - x(r) \rangle - \underbrace{\langle g(r), \bar{x} - x(r) \rangle}_{=(1-r)\langle g(r), \bar{x} - x_t \rangle \geq 0} \leq C,$$

meaning that the vectors $g(r) = \omega'_s(x(r))$ remain bounded as $r \rightarrow +0$. It follows that ω_s (and then ω) does have a subgradient at $x_t = x(0)$ (you can take as such a subgradient every limiting, as $r \rightarrow +0$, point of the curve $g(r)$). Recalling the smoothness properties of ω , we conclude that $\omega'_s(x_t)$ is well defined. With this in mind, (5.4.4) is proved by exactly the same argument as we have used to establish Fact 5.3.1.

5.4.2 Convergence Analysis

Let us define s -th gap as the quantity

$$\epsilon_s = f^s - f_s$$

By its origin, the gap is nonnegative and nonincreasing in s ; besides this, it clearly is an upper bound on the inaccuracy, in terms of the objective, of the approximate solution z^s we have at our disposal at the beginning of phase s .

The convergence and the complexity properties of the BM algorithm are given by the following statement.

Theorem 5.4.1 (i) *The number N_s of oracle calls at a phase s can be bounded from above as follows:*

$$N_s \leq \frac{5\Omega_s^2 L_{\|\cdot\|}^2(f)}{\theta^2(1-\lambda)^2\epsilon_s^2}, \quad \Omega_s = \sqrt{2 \max_{y \in X} [\omega(y) - \omega(c_s) - \langle \omega'(c_s), y - c_s \rangle]}. \quad (5.4.7)$$

(ii) *Let out policy for updating prox-centers ensure that*

$$\Omega_s \leq \hat{\Omega} < \infty, \quad s = 1, 2, \dots \quad (5.4.8)$$

Then for every $\epsilon > 0$ the total number of oracle calls before the phase s with $\epsilon_s \leq \epsilon$ is started (i.e., before ϵ -solution to the problem is built) does not exceed

$$N(\epsilon) = c(\theta, \lambda) \frac{\widehat{\Omega}^2 L_{\|\cdot\|}^2(f)}{\epsilon^2} \quad (5.4.9)$$

with an appropriate $c(\theta, \lambda)$ depending solely and continuously on $\theta, \lambda \in (0, 1)$.

Proof. (i): Assume that phase s did not terminate in course of N steps. Observe that then

$$\|x_t - x_{t-1}\| \geq \frac{\theta(1-\lambda)\epsilon_s}{L_{\|\cdot\|}(f)}, \quad 1 \leq t \leq N. \quad (5.4.10)$$

Indeed, we have $g_{t-1}(x_t) \leq \ell_s$ by construction of x_t and $g_{t-1}(x_{t-1}) = f(x_{t-1}) > \ell_s + \theta(f^s - \ell_s)$, since otherwise the phase would be terminated at the step $t-1$. It follows that $g_{t-1}(x_{t-1}) - g_{t-1}(x_t) > \theta(f^s - \ell_s) = \theta(1-\lambda)\epsilon_s$. Taking into account that $g_{t-1}(\cdot)$, due to its origin, is Lipschitz continuous w.r.t. $\|\cdot\|$ with constant $L_{\|\cdot\|}(f)$, (5.4.10) follows.

Now observe that x_{t-1} is the minimizer of ω_s on X_{t-1} by (5.4.2.a_{t-1}), and the latter set, by construction, contains x_t , whence $\langle x_t - x_{t-1}, \omega'_s(x_{t-1}) \rangle \geq 0$ (see (5.4.4)). Applying (5.4.1), we get

$$\omega_s(x_t) \geq \omega_s(x_{t-1}) + \frac{1}{2} \left(\frac{\theta(1-\lambda)\epsilon_s}{L_{\|\cdot\|}(f)} \right)^2, \quad t = 1, \dots, N,$$

whence

$$\omega_s(x_N) - \omega_s(x_0) \geq \frac{N}{2} \left(\frac{\theta(1-\lambda)\epsilon_s}{L_{\|\cdot\|}(f)} \right)^2.$$

The latter relation, due to the evident inequality $\max_X \omega_s(x) - \min_X \omega_s \leq \frac{\Omega_s^2}{2}$ (readily given by the definition of Ω_s and ω_s) implies that

$$N \leq \frac{\Omega_s^2 L_{\|\cdot\|}^2(f)}{\theta^2(1-\lambda)^2 \epsilon_s^2}.$$

Recalling the origin of N , we conclude that

$$N_s \leq \frac{\Omega_s^2 L_{\|\cdot\|}^2(f)}{\theta^2(1-\lambda)^2 \epsilon_s^2} + 1.$$

All we need in order to get from this inequality the required relation (5.4.7) is to demonstrate that

$$\frac{\Omega_s^2 L_{\|\cdot\|}^2(f)}{\theta^2(1-\lambda)^2 \epsilon_s^2} \geq \frac{1}{4}, \quad (5.4.11)$$

which is immediate: let $R = \max_{x \in X} \|x - c_1\|$. We have $\epsilon_s \leq \epsilon_1 = f(c_1) - \min_{x \in X} [f(c_1) + (x - c_1)^T f'(c_1)] \leq RL_{\|\cdot\|}(f)$, where the concluding inequality is given by the fact that $\|f'(c_1)\|_* \leq L_{\|\cdot\|}(f)$. On the other hand, by the definition of Ω_s we have $\frac{\Omega_s^2}{2} \geq \max_{y \in X} [\omega(y) - \langle \omega'(c_s), y - c_s \rangle - \omega(c_s)]$ and $\omega(y) - \langle \omega'(c_s), y - c_s \rangle - \omega(c_s) \geq \frac{1}{2} \|y - c_s\|^2$, whence $\|y - c_s\| \leq \Omega_s$ for all $y \in X$ (cf. (5.3.7)), and thus $R \leq 2\Omega_s$, whence $\epsilon_s \leq L_{\|\cdot\|}(f)R \leq 2L_{\|\cdot\|}(f)\Omega_s$, as required. (i) is proved.

(ii): Assume that $\epsilon_s > \epsilon$ at the phases $s = 1, 2, \dots, S$, and let us bound from above the total number of oracle calls at these S phases. Observe, first, that the two subsequent gaps $\epsilon_s, \epsilon_{s+1}$ are linked by the relation

$$\epsilon_{s+1} \leq \gamma \epsilon_s, \quad \gamma = \gamma(\theta, \lambda) \max[1 - \theta\lambda, 1 - (1 - \theta)(1 - \lambda)] < 1. \quad (5.4.12)$$

Indeed, it is possible that the phase s was terminated according to the rule 2a; in this case

$$\epsilon_{s+1} = f^{s+1} - f_{s+1} \leq f^s - [\ell_s - \theta(\ell_s - f_s)] = (1 - \theta\lambda)\epsilon_s,$$

as required in (5.4.12). The only other possibility is that the phase s was terminated when the relation (5.4.5) took place. In this case,

$$\epsilon_{s+1} = f^{s+1} - f_{s+1} \leq f^{s+1} - f_s \leq \ell_s + \theta(f^s - \ell_s) - f_s = \lambda\epsilon_s + \theta(1 - \lambda)\epsilon_s = (1 - (1 - \theta)(1 - \lambda))\epsilon_s,$$

and we again arrive at (5.4.12).

From (5.4.12) it follows that $\epsilon_s \geq \epsilon\gamma^{s-S}$, $s = 1, \dots, S$, since $\epsilon_S > \epsilon$ by the origin of S . We now have

$$\begin{aligned} \sum_{s=1}^S N_s &\leq \sum_{s=1}^S \frac{5\Omega_s^2 L_{\|\cdot\|}(f)}{\theta^2(1-\lambda)^2\epsilon_s^2} \leq \sum_{s=1}^S \frac{5\widehat{\Omega}^2 L_{\|\cdot\|}(f)\gamma^{2(S-s)}}{\theta^2(1-\lambda)^2\epsilon^2} \leq \frac{5\widehat{\Omega}^2 L_{\|\cdot\|}(f)}{\theta^2(1-\lambda)^2\epsilon^2} \sum_{t=0}^{\infty} \gamma^{2t} \\ &\equiv \underbrace{\left[\frac{5}{\theta^2(1-\lambda)^2(1-\gamma^2)} \right]}_{c(\theta,\lambda)} \frac{\widehat{\Omega}^2 L_{\|\cdot\|}(f)}{\epsilon^2} \end{aligned}$$

and (5.4.9) follows. ■

5.4.3 Implementation issues

5.4.3.1 Solving auxiliary problems (L_t) , (P_t)

As far as implementation of the BM algorithm is concerned, the major issue is *how to solve efficiently the auxiliary problems (L_t) , (P_t)* . Formally, these problems are of the same design dimension as the problem of interest; what do we gain when reducing the solution of a *single* large-scale problem (CP) to a long series of auxiliary problems of the same dimension? To answer this crucial question, observe first that we have control on the complexity of the domain X_t which, up to a single linear constraint, is the feasible domain of (L_t) , (P_t) . Indeed, assume that X_{t-1} is a part of X given by a finite list of linear inequalities. Then both sets \underline{X}_t and \overline{X}_t in (5.4.6) also are cut off X by finitely many linear inequalities, so that we may enforce X_t to be cut off X by finitely many linear inequalities as well. Moreover, we have full control of the number of inequalities in the list. For example,

- A. Setting all the time $X_t = \overline{X}_t$, we ensure that X_t is cut off X by a *single* linear inequality;
- B. Setting all the time $X_t = \underline{X}_t$, we ensure that X_t is cut off X by t linear inequalities (so that the larger is t , the “more complicated” is the description of X_t);
- C. We can choose something in-between the above extremes. For example, assume that we have chosen certain m and are ready to work with X_t ’s cut off X by at most m linear inequalities. In this case, we could use the policy B. at the initial steps of a phase, until the number of linear inequalities in the description of X_{t-1} reaches the maximal allowed value m . At the step t , we are supposed to choose X_t in-between the two sets

$$\underline{X}_t = \{x \in X : h_1(x) \leq 0, \dots, h_m(x) \leq 0, h_{m+1}(x) \leq 0\}, \quad \overline{X}_t = \{x \in X : h_{m+2}(x) \leq 0\},$$

where

- the linear inequalities $h_1(x) \leq 0, \dots, h_m(x) \leq 0$ cut X_{t-1} off X ;
- the linear inequality $h_{m+1}(x) \leq 0$ is, in our old notation, the inequality $g_{t-1}(x) \leq \ell_s$;
- the linear inequality $h_{m+2}(x) \leq 0$ is, in our old notation, the inequality $\langle x - x_t, \omega'_s(x_t) \rangle \geq 0$.

Now we can form a list of m linear inequalities as follows:

- we build $m - 1$ linear inequalities $e_i(x) \leq 0$ by aggregating the $m + 1$ inequalities $h_j(x) \leq 0$, $j = 1, \dots, m + 1$, so that every one of the e -inequalities is a convex combination of the h -ones (the coefficients of these convex combinations can be whatever we want);
- we set $X_t = \{x \in X : e_i(x) \leq 0, i = 1, \dots, m - 1, h_{m+2}(x) \leq 0\}$.

It is immediately seen that with this approach we ensure (5.4.6), on one hand, and that X_t is cut off X by at most m inequalities. And of course we can proceed in the same fashion.

The bottom line is: we always can ensure that X_{t-1} is cut off X by at most m linear inequalities $h_j(x) \leq 0$, $j = 1, \dots, m$, where $m \geq 1$ is (any) desirable bound. Consequently, we may assume that the feasible set of (P_{t-1}) is cut off X by $m + 1$ linear inequalities $h_j(x) \leq 0$, $j = 1, \dots, m + 1$. The crucial point is that with this approach, we can reduce (L_{t-1}) , (P_{t-1}) to convex programs with at most $m + 1$ decision variables. Indeed, replacing $\mathbf{R}^n \supset X$ with the affine span of X , we may assume the X has a nonempty interior. By (o_{t-1}) this is also the case with the feasible domain X_{t-1} of the problem

$$\text{Opt} = \min_x \{g_{t-1}(x) : x \in X_{t-1} = \{x \in X : h_j(x) \leq 0, 1 \leq j \leq m\}\} \quad (L_{t-1}),$$

implying that the linear constraints $h_j(x) \leq 0$ (which we w.l.o.g. assume to be nonconstant) satisfy the Slater condition on $\text{int } X$: there exists $\bar{x} \in \text{int } X$ such that $h_j(\bar{x}) \leq 0$, $j \leq m$. Applying Convex Programming Duality Theorem (Theorem D.2.2), we get

$$\text{Opt} = \max_{\lambda \geq 0} G(\lambda), \quad G(\lambda) = \min_{x \in X} [g_{t-1}(x) + \sum_{j=1}^m \lambda_j h_j(x)]. \quad (\text{D})$$

Now, assuming that prox-mapping associated with $\omega(\cdot)$ is easy to compute, it is equally easy to minimize an affine function over X (why?), so that $G(\lambda)$ is easy to compute:

$$G(\lambda) = g_{t-1}(x_\lambda) + \sum_{j=1}^m \lambda_j h_j(x_\lambda), \quad x_\lambda \in \operatorname{Argmin}_{x \in X} [g_{t-1}(x) + \sum_{j=1}^m \lambda_j h_j(x)].$$

By construction, $G(\lambda)$ is a well defined concave function; its supergradient is given by

$$G'(\lambda) = [h_1(x_\lambda); \dots; h_m(x_\lambda)]$$

and thus is as readily available as the value $G(\lambda)$ of G . It follows that we can compute Opt (which is all we need as far as the auxiliary problem (L_{t-1}) is concerned) by solving the m -dimensional convex program (D) by a whatever rapidly converging first order method¹¹.

The situation with the second auxiliary problem

$$\min_x \{\omega_s(x) : x \in X_{t-1}, g_{t-1}(x) \leq \ell_s\} \quad (P_{t-1}).$$

is similar. Here again we are interested to solve the problem only in the case when its feasible domain is full-dimensional (recall that we w.l.o.g. have assumed X to be full-dimensional), and the constraints cutting the feasible domain off X are linear, so that we can write (P_{t-1}) down as the problem

$$\min_x \{\omega_s(x) : h_j(x) \leq 0, 1 \leq j \leq m + 1\}$$

¹¹e.g., by the Ellipsoid algorithm, provided m is in range of tens.

with linear constraints $h_j(x) \leq 0$, $1 \leq j \leq m+1$, satisfying Slater condition on $\text{int } X$. As before, we can pass to the dual problem

$$\max_{\lambda \geq 0} H(\lambda), \quad H(\lambda) = \min_{x \in X} \left[\omega_s(x) + \sum_{j=1}^{m+1} \lambda_j h_j(x) \right]. \quad (\text{D}_{t-1})$$

Here again, the concave and well defined dual objective $H(\lambda)$ can be equipped with a cheap First Order oracle (provided, as above, that the prox-mapping associated with $\omega(\cdot)$ is easy to compute), so that we can reduce the auxiliary problem (P_{t-1}) to solving a low-dimensional (provided m is not large) black-box-represented convex program by a first order method. Note that now we want to build a high-accuracy approximation to the *optimal solution* of the problem of actual interest (P_{t-1}) rather than to approximate well its optimal value, and a high-accuracy solution to the Lagrange dual of a convex program not always allows to recover a good solution to the program itself. Fortunately, in our case the objective of the problem of actual interest (P_{t-1}) is strongly convex, and in this case a high accuracy solution λ to the dual problem (D_{t-1}) , the accuracy being measured in terms of the dual objective H , does produce a high-accuracy solution

$$x_\lambda = \operatorname{argmin}_X \left[\omega_s(x) + \sum_{j=1}^{m+1} \lambda_j h_j(x) \right]$$

to (P_{t-1}) , the accuracy being measured in terms of the distance to the precise solution x_t to the latter problem.

The bottom line is that when the “memory depth” m (which is fully controlled by us!) is not too big and the prox-mapping associated with ω is easy to compute, the auxiliary problems arising in the BM algorithm can be solved, at a relatively low computational cost, via Lagrange duality combined with good high-accuracy first order algorithm for solving black-box-represented low dimensional convex programs.

5.4.3.2 Updating prox-centers

The complexity results stated in Theorem 5.4.1 depend somehow on how we update the prox-centers — what we need is a uniform in s above boundedness of the quantities Ω_s , and the more is their common bound, the larger is the complexity. A reasonable policy here heavily depends on whether the DGF is continuously differentiable on the entire X (as it is the case with Ball and the Simplex/Nuclear norm/Spectahedron setups), or it is not so (as is the case with Entropy setup). When ω is continuously differentiable on the entire X , we have $X^o = X$ and

$$\Omega(x) := \sqrt{2 \max_{y \in X} [\omega(y) - \omega(x) - \langle \omega'(x), y - x \rangle]} \leq \tilde{\Omega} := \max_{x \in X} \Omega(x) < \infty.$$

Thus, we can choose $c_s \in X$ as we want, at the price of setting $\hat{\Omega} = \tilde{\Omega}$. Note that when X is either a Euclidean ball/box and Ball setup is used, or is a $\ell_1\ell_2$ ball, or nuclear norm ball, or spectahedron, and the respective of ℓ_1/ℓ_2 /Nuclear norm/Spectahedron setups is used, the outlined price is not too high (since on a closest inspection, the quantities $\tilde{\Omega}$ in these cases are, up to moderate dimension independent (or nearly so) factors, the best possible). Practical experience says that in the situations in question, when there is a full freedom in choosing prox-centers, a reasonable policy is to use as c_s the best, in terms of the values of f , search point generated so far.

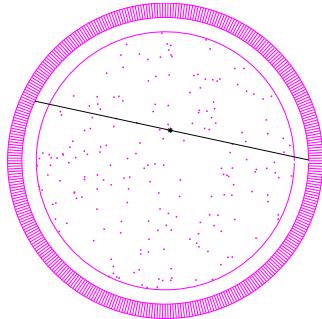


Figure 5.1. Ring with 360 detectors, field of view and a line of response

When ω is not continuously differentiable, it might be dangerous to place the prox-centers close to the relative boundary of X , since this can result in large values of Ω_s and thus of $\widehat{\Omega}$. The theoretically “most safe” policy here is to keep $c_s = x_\omega$ for all s , thus ensuring that $\widehat{\Omega} = \Omega$.

5.4.3.3 Accumulating information

The set X_t summarizes, in a sense, all the information on f we have accumulated so far and intend to use in the sequel. Relation (5.4.6) allows for a tradeoff between the quality (and the volume) of this information and the computational effort required to solve the auxiliary problems (P_{t-1}). With no restrictions on this effort, the most promising policy for updating X_t ’s would be to set $X_t = \underline{X}_{t-1}$ (“collecting information with no compression of it”). With this policy the BM algorithm with the ball setup is basically identical to the *Prox-Level Algorithm* of Lemarechal, Nemirovski and Nesterov [18]; the “truncated memory” version of the latter method (that is, the generic BM algorithm with ball setup) was proposed by Kiwiel [17]. Aside of theoretical complexity bounds similar to (5.3.37), most of bundle methods (in particular, the Prox-Level one) share the following experimentally observed property: *the practical performance of the algorithm is in full accordance with the complexity bound (5.1.2): every n steps reduce the inaccuracy by at least an absolute constant factor* (something like 3). This property is very attractive in moderate dimensions, where we indeed are capable to carry out several times the dimension number of steps.

5.4.4 Illustration: PET Image Reconstruction problem by MD and BM

To get an impression of the practical performance of the BM method, let us look at numerical results related to the 2D version of the PET Image Reconstruction problem.

The model. We process simulated measurements as if they were registered by a ring of 360 detectors, the inner radius of the ring being 1 (Fig. 5.1). The field of view is a concentric circle of the radius 0.9, and it is covered by the 129×129 rectangular grid. The grid partitions the field of view into 10,471 pixels, and we act as if tracer’s density was constant in every pixel. Thus, the design dimension of the problem (PET’) we are interested to solve is “just” $n = 10471$.

The number of bins (i.e., number m of log-terms in the objective of (PET’)) is 39784, while the number of nonzeros among q_{ij} is 3,746,832.

The true image is “black and white” (the density in every pixel is either 1 or 0). The measurement time (which is responsible for the level of noise in the measurements) is mimicked as follows: we model the measurements according to the Poisson model as if during the period of measurements the expected number of positrons emitted by a single pixel with unit density was a given number M .

The algorithm we are using to solve (PET') is the plain BM method with the simplex setup and the sets X_t cut off $X = \Delta_n$ by just one linear inequality:

$$X_t = \{x \in \Delta_n : (x - x_t)^T \nabla \omega_s(x_t) \geq 0\}.$$

The parameters λ, θ of the algorithm were chosen as

$$\lambda = 0.95, \quad \theta = 0.5.$$

The approximate solution reported by the algorithm at a step is the best found so far search point (the one with the best value of the objective we have seen to the moment).

The results of two sample runs we are about to present are not that bad.

Experiment 1: Noiseless measurements. The evolution of the best, in terms of the objective, solutions x^t found in course of the first t calls to the oracle is displayed at Fig. 5.2 (on pictures, brighter areas correspond to higher density of the tracer). The numbers are as follows. With the noiseless measurements, we know in advance the optimal value in (PET') – it is easily seen that without noises, the true image (which in our simulated experiment we do know) is an optimal solution. In our problem, this optimal value equals to 2.8167; the best value of the objective found in 111 oracle calls is 2.8171 (optimality gap 4.e-4). The progress in accuracy is plotted on Fig. 5.3. We have built totally 111 search points, and the entire computation took 18'51" on a 350 MHz Pentium II laptop with 96 MB RAM ¹².

Experiment 2: Noisy measurements (40 LOR's per pixel with unit density, totally 63,092 LOR's registered). The pictures are presented at Fig. 5.4. Here are the numbers. With noisy measurements, we have no a priori knowledge of the true optimal value in (PET'); in simulated experiments, a kind of orientation is given by the value of the objective at the true image (which is hopefully a close to f_* upper bound on f_*). In our experiment, this bound equals to -0.8827. The best value of the objective found in 115 oracle calls is -0.8976 (which is less than the objective at the true image in fact, the algorithm went below the value of f at the true image already after 35 oracle calls). The upper bound on the optimality gap at termination is 9.7e-4. The progress in accuracy is plotted on Fig. 5.5. We have built totally 115 search points; the entire computation took 20'41".

3D PET. As it happened, the BT algorithm was not tested on actual clinical 3D PET data. However, in late 1990's we were participating in a EU project on 3D PET Image Reconstruction; in this project, among other things, we did apply to the actual 3D clinical data an algorithm which, up to minor details, is pretty close to the MD with Entropy setup as applied to convex minimization over the standard simplex (details can be found in [2]). In the sequel, with slight

¹²Do not be surprised by the obsolete hardware – the computations were carried out in 2002

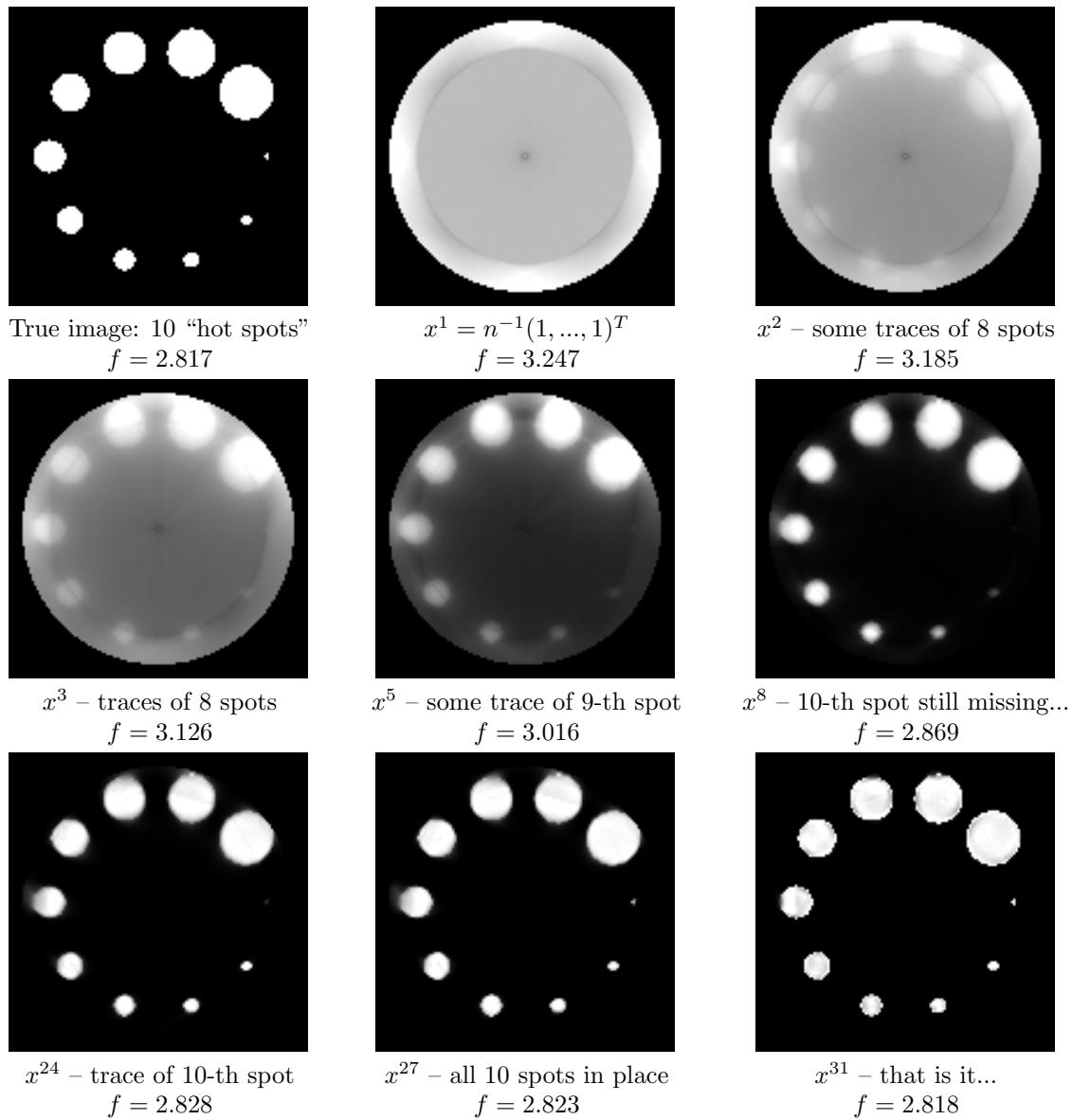


Figure 5.2. Reconstruction from noiseless measurements

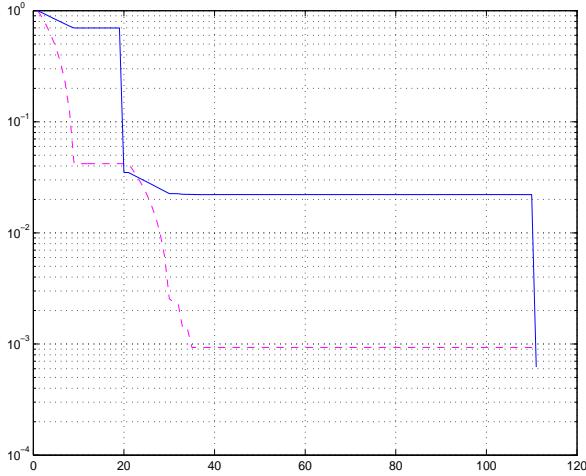


Figure 5.3. Progress in accuracy, noiseless measurements.

- solid line: Relative gap $\frac{\text{Gap}(t)}{\text{Gap}(1)}$ vs. step number t ; $\text{Gap}(t)$ is the difference between the best found so far value $f(x^t)$ of f and the current lower bound on f_* .
In 111 steps, the gap was reduced by factor > 1600
- dashed line: Progress in accuracy $\frac{f(x^t) - f_*}{f(x^1) - f_*}$ vs. step number t
In 111 steps, the accuracy was improved by factor > 1080

abuse of terminology, we refer to this algorithm as to MD, and present some experimental data on its practical performance in order to demonstrate that simple optimization techniques indeed have chances when applied to really huge convex programs. We restrict ourselves with a single numerical example – real clinical brain study carried out on a very powerful PET scanner. In the corresponding problem (PET'), there are $n = 2,763,635$ design variables (this is the number of voxels in the field of view) and about 25,000,000 log-terms in the objective. The reconstruction was carried out on the INTEL Marlinspike Windows NT Workstation (500 MHz 1Mb Cache INTEL Pentium III Xeon processor, 2GB RAM; this hardware, completely obsolete now, was state-of-the-art one in late 1990s). A single call to the First Order oracle (a single computation of the value and the subgradient of f) took about 90 min. Pictures of clinically acceptable quality were obtained after just four calls to the oracle (as it was the case with other sets of PET data); for research purposes, we carried out 6 additional steps of the algorithm.

The pictures presented on Fig. 5.6 are *slices* – 2D cross-sections – of the reconstructed 3D image. Two series of pictures shown on Fig. 5.6 correspond to two different versions, plain and “ordered subsets” one (MD and OSMD, respectively), of the method, see [2] for details. Relevant numbers are presented in Table 5.1. The best known lower bound on the optimal value in the problem is -2.050; MD and OSMD decrease in 10 oracle calls the objective from its initial value -1.436 to the values -2.009 and -2.016, respectively (optimality gaps 4.e-2 and 3.5e-2) and reduce the initial inaccuracy in terms of the objective by factors 15.3 (MD) and 17.5 (OSMD).

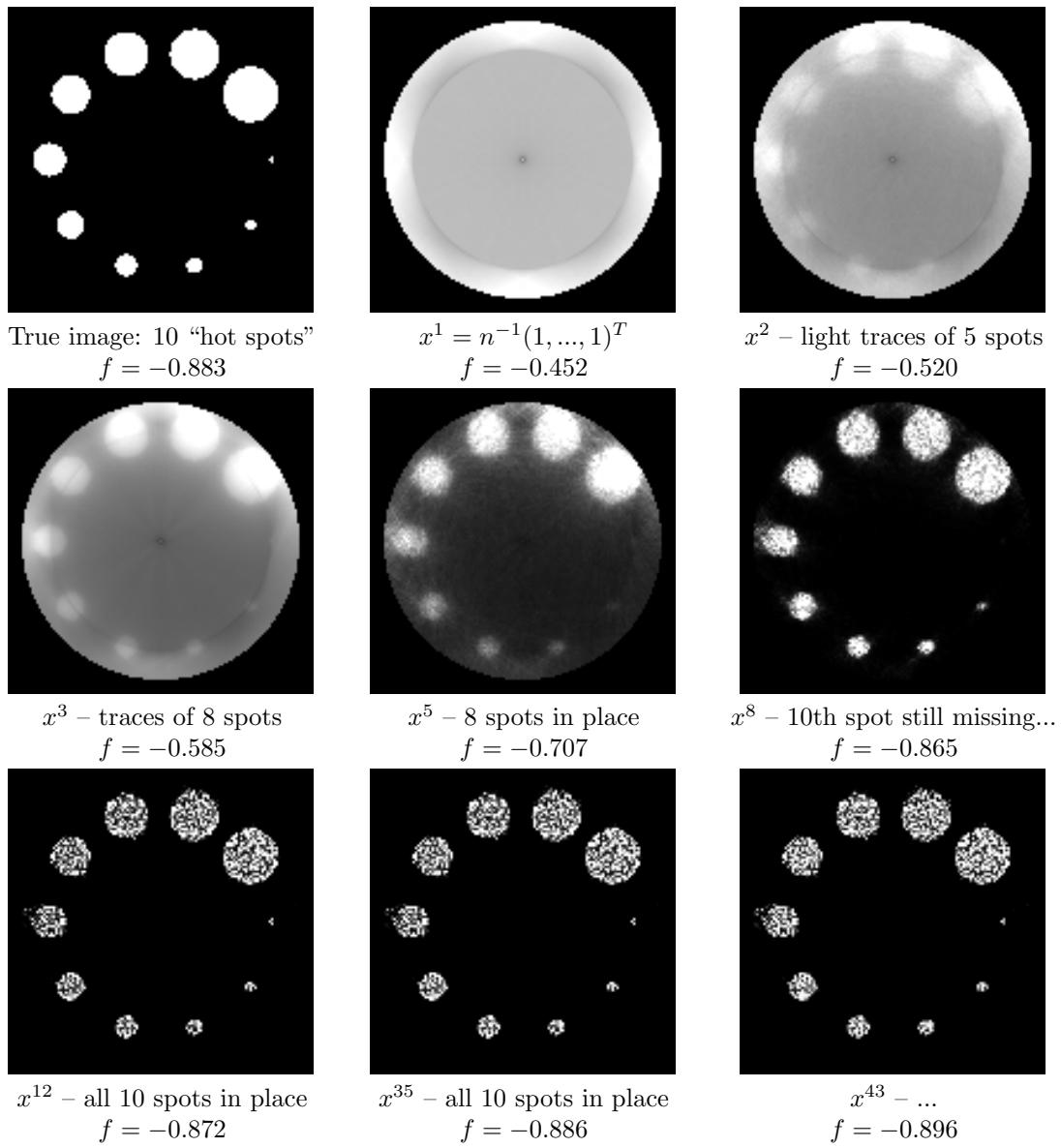


Figure 5.4. Reconstruction from noisy measurements

step #	MD	OSMD	step #	MD	OSMD
1	-1.463	-1.463	6	-1.987	-2.015
2	-1.725	-1.848	7	-1.997	-2.016
3	-1.867	-2.001	8	-2.008	-2.016
4	-1.951	-2.015	9	-2.008	-2.016
5	-1.987	-2.015	10	-2.009	-2.016

Table 5.1. Performance of MD and OSMD in Brain study

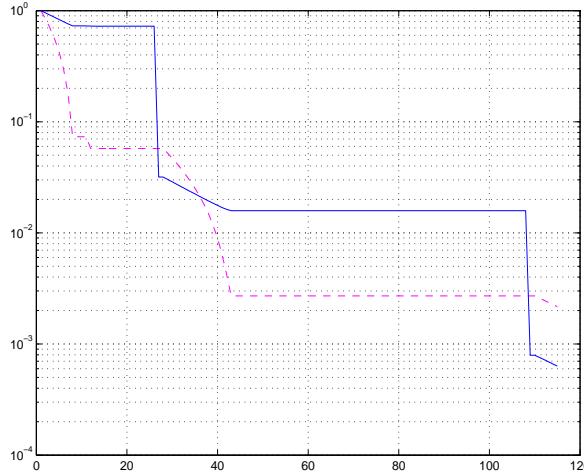


Figure 5.5. Progress in accuracy, noisy measurements.

- solid line: Relative gap $\frac{|Gap(t)|}{|Gap(1)|}$ vs. step number t
In 115 steps, the gap was reduced by factor 1580
- dashed line: Progress in accuracy $\frac{f(x^t) - f}{f(x^1) - \underline{f}}$ vs. step number t
(\underline{f} is the last lower bound on f_* built in the run)
In 115 steps, the accuracy was improved by factor > 460

5.5 Fast First Order algorithms for Convex Minimization and Convex-Concave Saddle Points

5.5.1 Fast Gradient Methods for Smooth Composite minimization

The fact that problem (CP) with smooth convex objective can be solved by a gradient-type algorithm at the rate $O(1/T^2)$ was discovered by Yu. Nesterov in 1983 (celebrated Nesterov's *optimal algorithm for large-scale smooth convex optimization* [25]). A significant more recent development here, again due to Yu. Nesterov [30] is extending of the $O(1/T^2)$ -converging algorithm from the case of problem (CP) with smooth objective (smoothness is a rare commodity!) to the so called *composite case* where the objective in (CP) is the sum of a smooth convex component and a nonsmooth "easy to handle" convex component. We are about to present the related results of Yu. Nesterov; our exposition follows [31] (modulo minor modifications aimed at handling inexact proximal mappings).

5.5.1.1 Problem formulation

The general problem of composite minimization is

$$\phi_* = \min_{x \in X} \{\phi(x) := f(x) + \Psi(x)\}, \quad (5.5.1)$$

where X is a closed convex set in a Euclidean space E equipped with norm $\|\cdot\|$ (not necessarily the Euclidean one), $\Psi(x) : X \rightarrow \mathbf{R} \cup \{+\infty\}$ is a lower semicontinuous convex function which is finite on the relative interior of X , and $f : X \rightarrow \mathbf{R}$ is a convex function. In this section, we

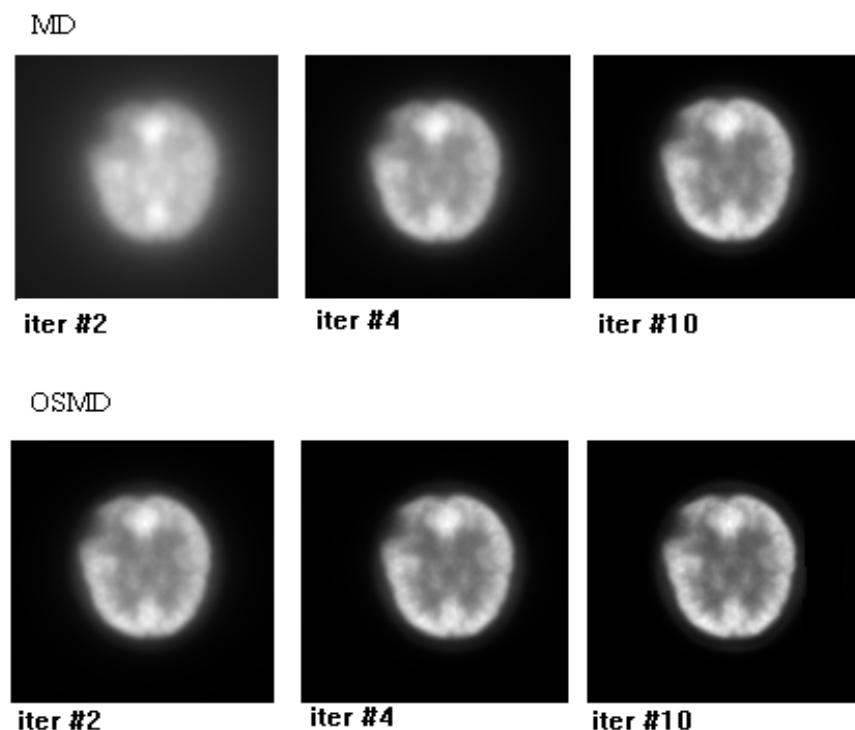


Figure 5.6. Brain, near-mid slice of the reconstructions.
[the top-left missing part is the area affected by the Alzheimer disease]

focus on the *smooth composite case*, where f has Lipschitz continuous gradient:

$$\|\nabla f(x) - \nabla f(y)\|_* \leq L_f \|x - y\|, \quad x, y \in X; \quad (5.5.2)$$

here, as always, $\|\cdot\|_*$ is the norm conjugate to $\|\cdot\|$. Condition (5.5.2) ensures that

$$f(y) - f(x) - \langle \nabla f(x), y - x \rangle \leq \frac{L_f}{2} \|x - y\|^2 \quad \forall x, y \in X, \quad (5.5.3)$$

Note that we do *not* assume here that X is bounded; however, we do assume from now on that (5.5.1) is solvable, and denote by x_* an optimal solution to the problem.

5.5.1.2 Composite prox-mapping

We assume that X is equipped with DGF $\omega(x)$ compatible with $\|\cdot\|$. As always, we set $X^o = \{x \in X : \partial_X \omega(x) \neq \emptyset\}$,

$$V_x(y) = \omega(y) - \omega(x) - \langle \omega'(x), y - x \rangle \quad [x \in X^o, y \in X]$$

and denote by x_ω the minimizer of $\omega(\cdot)$ on X .

Given a convex lower semicontinuous function $h(x) : X \rightarrow \mathbf{R} \cup \{+\infty\}$ which is finite on the relative interior of X , we know from Lemma 5.6.1 that the optimization problem

$$\text{Opt} = \min_{x \in X} \{h(x) + \omega(x)\}$$

is solvable with the unique solution $z = \operatorname{argmin}_{x \in X} [h(x) + \omega(x)]$ belonging to X^o and fully characterized by the property

$$z \in X^o \& \{\exists h'(z) \in \partial_X h(z) : \langle h'(z) + \omega'(z), w - z \rangle \geq 0 \quad \forall w \in X\}.$$

Given $\epsilon \geq 0$ and h as above, let us set

$$\operatorname{argmin}_{x \in X}^\epsilon \{h(x) + \omega(x)\} = \{z \in X^o : \exists h'(z) \in \partial_X h(z) : \langle h'(z) + \omega'(z), w - z \rangle \geq -\epsilon \quad \forall w \in X\}$$

Note that for h as above, the set $\operatorname{argmin}_{x \in X}^\epsilon \{h(x) + \omega(x)\}$ is nonempty, since it clearly contains the point $\operatorname{argmin}_{x \in X} \{h(x) + \omega(x)\}$.

The algorithm we are about to develop requires the ability to find, given $\xi \in E$ and $\alpha \geq 0$, a point from the set

$$\operatorname{argmin}_{x \in X}^\epsilon \{\langle \xi, x \rangle + \alpha \Psi(x) + \omega(x)\}$$

where $\epsilon \geq 0$ is construction's parameter¹³; we refer to such a computation as to *computing within accuracy ϵ a composite prox-mapping*, ξ , α being the arguments of the mapping. The “ideal” situation here is when the *composite prox-mapping*

$$(\xi, \alpha) \mapsto \operatorname{argmin}_{x \in X} \{\langle \xi, x \rangle + \alpha \Psi(x) + \omega(x)\}$$

is easy to compute either by an explicit formula, or by a cheap computational procedure, as is the case in the following instructive example.

¹³In the usual Composite FGM, $\epsilon = 0$; we allow for $\epsilon > 0$ in order to incorporate eventually the case where X is given by Linear Minimization Oracle, see Corollary 5.5.2.

Example: LASSO. An instructive example of a composite minimization problem where the composite prox-mapping indeed is easy to compute is the LASSO problem

$$\min_{x \in E} \left\{ \lambda \|x\|_E + \|A(x) - b\|_2^2 \right\}, \quad (5.5.4)$$

where E is a Euclidean space, $\|\cdot\|_E$ is a norm on E , $x \mapsto A(x) : E \rightarrow \mathbf{R}^m$ is a linear mapping, $b \in \mathbf{R}^m$, and $\lambda > 0$ is a penalty coefficient. To represent the LASSO problem in the form of (5.5.1), it suffices to set

$$f(x) = \|A(x) - b\|_2^2, \quad \Psi(x) = \|x\|_E, \quad X = E.$$

The most important cases of the LASSO problem are

A. Sparse recovery: $E = \mathbf{R}^n$, $\|\cdot\| = \|\cdot\|_1$ or, more generally, $E = \mathbf{R}^{k_1} \times \dots \times \mathbf{R}^{k_n}$ and $\|\cdot\|$ is the associated block ℓ_1 norm. In this case, thinking of x as of the unknown input to a linear system, and of b as of observed (perhaps, corrupted by noise) output, (5.5.4) is used to carry out a trade off between “model fitting term” $\|A(x) - b\|_2^2$ and the (block) sparsity of x : usually, the larger is λ , the less nonzero entries (in the $\|\cdot\|_1$ -case) or nonzero blocks (in the block ℓ_1 case) has the optimal solution to (5.5.4), and the larger is the model fitting term.

Note that in the case in question, computing composite prox-mapping is really easy, provided that the proximal setup in use is either

- (a) the Euclidean setup, or
- (b) the ℓ_1/ℓ_2 setup with DGF (5.3.12).

Indeed, in both cases computing composite prox-mapping reduces to solving the optimization problem of the form

$$\min_{x^j \in \mathbf{R}^{k_j}, 1 \leq j \leq n} \left\{ \sum_j [\alpha \|x^j\|_2 + [b^j]^T x^j] + \left(\sum_j \|x^j\|_2^p \right)^{2/p} \right\}, \quad (5.5.5)$$

where $\alpha > 0$ and $1 < p \leq 2$. This problem has a closed form solution (find it!) which can be computed in $O(\dim E)$ a.o.

B. Low rank recovery: $E = \mathbf{M}^{\mu\nu}$, $\|\cdot\| = \|\cdot\|_{\text{nuc}}$ is the nuclear norm. The rationale behind this setting is similar to the one for sparse recovery, but now the trade off is between the model fitting and the *rank* of the optimal solution (which is a block-diagonal matrix) to (5.5.4).

In the case in question, for both the Euclidean and the Nuclear norm proximal setups (in the latter case, one should take $X = E$ and use the DGF (5.3.17)), so that computing composite prox-mapping reduces to solving the optimization problem

$$\min_{y = \text{Diag}\{y^1, \dots, y^n\} \in \mathbf{M}^{\mu\nu}} \left\{ \sum_{j=1}^n \text{Tr}(y^j [b^j]^T) + \alpha \|\sigma(y)\|_1 + \left(\sum_{i=1}^m \sigma_i^p(y) \right)^{2/p} \right\}, \quad m = \sum_j \mu_j, \quad (5.5.6)$$

where $b = \text{Diag}\{b^1, \dots, b^n\} \in \mathbf{M}^{\mu\nu}$, $\alpha > 0$ and $p \in (1, 2]$. Same as in the case of the usual prox-mapping associated with the Nuclear norm setup (see section 5.3.3), after computing the svd of b the problem reduces to the similar one with *diagonal* b^j 's: $b^j = [\text{Diag}\{\beta_1^j, \dots, \beta_{\mu_j}^j\}, 0_{\mu_j \times (\nu_j - \mu_j)}]$, where $\beta_k^j \geq 0$. By the same symmetry argument as in section 5.3.3, in the case of diagonal b^j , (5.5.6) has an optimal solution with diagonal y^j 's. Denoting s_k^j , $1 \leq k \leq j$, the diagonal entries

in the blocks y^j of this solution, (5.5.6) reduces to the problem

$$\min_{\{s_k^j : q \leq j \leq n, 1 \leq k \leq \mu_j\}} \left\{ \sum_{\substack{1 \leq j \leq n, \\ 1 \leq k \leq \mu_j}} [s_k^j \beta_k^j + \alpha |s_k^j|] + \left(\sum_{\substack{1 \leq j \leq n, \\ 1 \leq k \leq \mu_j}} |s_k^j|^p \right)^{2/p} \right\},$$

which is of exactly the same structure as (5.5.5) and thus admits a closed form solution which can be computed at the cost of $O(m)$ a.o. We see that in the case in question computing composite prox-mapping is no more involved than computing the usual prox-mapping and reduces, essentially, to finding the svd of a matrix from $\mathbf{M}^{\mu\nu}$.

5.5.1.3 Fast Composite Gradient minimization: Algorithm and Main Result

In order to simplify our presentation, we assume that the constant $L_f > 0$ appearing in (5.5.2) is known in advance.

The algorithm we are about to develop depends on a sequence $\{L_t\}_{t=0}^\infty$ of reals such that

$$0 < L_t \leq L_f, t = 0, 1, \dots \quad (5.5.7)$$

In order to ensure $O(1/t^2)$ convergence, the sequence should satisfy certain conditions which can be verified online; to meet these conditions, the sequence can be adjusted online. We shall discuss this issue in-depth later.

Let us fix two tolerances $\epsilon \geq 0$, $\bar{\epsilon} \geq 0$. The algorithm is as follows:

- **Initialization:** Set $y_0 = x_\omega := \operatorname{argmin}_{x \in X} \omega(x)$, $\psi_0(w) = V_{x_\omega}(w)$, and select $y_0^+ \in X$ such that $\phi(y_0^+) \leq \phi(y_0)$. Set $A_0 = 0$.
- **Step** $t = 0, 1, \dots$ Given $\psi_t(\cdot)$, y_t^+ and $A_t \geq 0$,

1. Compute

$$z_t \in \operatorname{argmin}_{w \in X}^\epsilon \psi_t(w). \quad (5.5.8)$$

2. Find the positive root a_{t+1} of the quadratic equation

$$L_t a_{t+1}^2 = A_t + a_{t+1} \quad \left[\Leftrightarrow a_{t+1} = \frac{1}{2L_t} + \sqrt{\frac{1}{4L_t^2} + \frac{A_t}{L_t}} \right] \quad (5.5.9)$$

and set

$$A_{t+1} = A_t + a_{t+1}, \quad \tau_t = \frac{a_{t+1}}{A_{t+1}}. \quad (5.5.10)$$

Note that $\tau_t \in (0, 1]$.

3. Set

$$x_{t+1} = \tau_t z_t + (1 - \tau_t) y_t^+. \quad (5.5.11)$$

and compute $f(x_{t+1})$, $\nabla f(x_{t+1})$.

4. Compute

$$\hat{x}_{t+1} \in \operatorname{argmin}_{w \in X}^{\bar{\epsilon}} [a_{t+1} [\langle \nabla f(x_{t+1}), w \rangle + \Psi(w)] + V_{z_t}(w)]. \quad (5.5.12)$$

5. Set

$$\begin{aligned} (a) \quad y_{t+1} &= \tau_t \hat{x}_{t+1} + (1 - \tau_t) y_t^+ \\ (b) \quad \psi_{t+1}(w) &= \psi_t(w) + a_{t+1} [f(x_{t+1}) + \langle \nabla f(x_{t+1}), w - x_{t+1} \rangle + \Psi(w)]. \end{aligned} \quad (5.5.13)$$

compute $f(y_{t+1})$, set

$$\delta_t = \left[\frac{1}{A_{t+1}} V_{z_t}(\hat{x}_{t+1}) + \langle \nabla f(x_{t+1}), y_{t+1} - x_{t+1} \rangle + f(x_{t+1}) \right] - f(y_{t+1}) \quad (5.5.14)$$

and select somehow $y_{t+1}^+ \in X$ in such a way that $\phi(y_{t+1}^+) \leq \phi(y_{t+1})$.

Step t is completed, and we pass to step $t + 1$.

The approximate solution generated by the method in course of t steps is y_{t+1}^+ .

In the sequel, we refer to the just described algorithm as to *Fast Composite Gradient Method*, nicknamed FGM.

Theorem 5.5.1 *FGM is well defined and ensures that $z_t \in X^o$, $x_{t+1}, \hat{x}_{t+1}, y_t, y_t^+ \in X$ for all $t \geq 0$. Assuming that (5.5.7) holds true and that $\delta_t \geq 0$ for all t (the latter definitely is the case when $L_t = L_f$ for all t), one has*

$$y_t, y_t^+ \in X, \phi(y_t^+) - \phi_* \leq \phi(y_t) - \phi_* \leq A_t^{-1} [V_{x_\omega}(x_*) + 2\epsilon t] \leq \frac{4L_f}{t^2} [V_{x_\omega}(x_*) + 2\epsilon t] \quad (5.5.15)$$

for $t = 1, 2, \dots$

A step of the algorithm, modulo updates $y_t \mapsto y_t^+$, reduces to computing the value and the gradient of f at a point, computing the value of f at another point and to computing within accuracy ϵ two prox mappings.

Comments. The $O(1/t^2)$ efficiency estimate (5.5.15) is conditional, the condition being that the quantities δ_t defined by (5.5.14) are nonnegative for all t and that $0 < L_t \leq L_f$ for all t . Invoking (5.5.3), we see that at every step t , setting $L_t = L_f$ ensures that $\delta_t \geq 0$. In other words, we can safely use $L_t \equiv L_f$ for all t . However, from the description of A_t 's it follows that the less are L_t 's, the larger are A_t 's and thus the better is the efficiency estimate (5.5.15). This observations suggests more “aggressive” policies aimed at operating with L_t 's much smaller than L_f . A typical policy of this type is as follows: at the beginning of step t , we have at our disposal a trial value $L_t^0 \leq L_f$ of L_t . We start with carrying out step t with L_t^0 in the role of L_t ; if we are lucky and this choice of L_t results in $\delta_t \geq 0$, we pass to step $t + 1$ and reduce the trial value of L , say, set $L_{t+1}^0 = L_t^0/2$. If with $L_t = L_t^0$ we get $\delta_t < 0$, we rerun step t with increased value of L_t , say, with $L_t = \min[2L_t^0, L_f]$ or even with $L_t = L_f$. If this new value of L_t still does not result in $\delta_t \geq 0$, we rerun step t with larger value of L_t , say, with $L_t = \min[4L_t^0, L_f]$, and proceed in this fashion until arriving at $L_t \leq L_f$ resulting in $\delta_t \geq 0$. We can maintain a whatever upper bound $k \geq 1$ on the number of trials at a step, just by setting $L_t = L_f$ at k -th trial. As far as the evolution of initial trial values L_t^0 of L_t in time is concerned, a reasonable “aggressive” policy could be “set L_{t+1}^0 to a fixed fraction (say, $1/2$) of the last (the one resulting in $\delta_t \geq 0$) value of L_t tested at step t .”

In spite of the fact that every trial requires some computational effort, the outlined on-line adjustment of L_t 's usually significantly outperforms the “safe” policy $L_t \equiv L_f$.

5.5.1.4 Proof of Theorem 5.5.1

1⁰. Observe that by construction

$$\psi_t(\cdot) = \omega(\cdot) + \ell_t(\cdot) + \alpha_t \Psi(\cdot) \quad (5.5.16)$$

with nonnegative α_t and affine $\ell_t(\cdot)$, whence, taking into account that $\tau_t \in (0, 1]$, the algorithm is well defined and maintains the inclusions $z_t \in X^o$, $x_t, \hat{x}_{t+1}, y_t, y_t^+ \in X$.

Besides this, (5.5.16), (5.5.12) show that computing z_t , same as computing \hat{x}_{t+1} , reduces to computing within accuracy ϵ composite prox-mappings, and the remaining computational effort at step t is dominated by the necessity to compute $f(x_{t+1})$, $\nabla f(x_{t+1})$, and $f(y_{t+1})$, as stated in Theorem.

2⁰. Observe that by construction

$$A_t = \sum_{i=1}^t a_i \quad (5.5.17)$$

(as always, the empty sum is 0), and

$$\frac{1}{2\tau_t^2 A_{t+1}} = \frac{A_{t+1}}{2a_{t+1}^2} = \frac{A_t + a_{t+1}}{2a_{t+1}^2} = \frac{L_t}{2}, \quad t \geq 0 \quad (5.5.18)$$

(see (5.5.9) and (5.5.10)). Observe also that the initialization rules, (5.5.17.a) and (5.5.13.b) imply that whenever $t \geq 1$, setting $\lambda_i^t = a_i/A_t$ we have

$$\begin{aligned} \psi_t(w) &= \sum_{i=1}^t a_i [\langle \nabla f(x_i), w - x_i \rangle + f(x_i) + \Psi(w)] + V_{x_\omega}(w) \\ &= A_t \sum_{i=1}^t \lambda_i^t [\langle \nabla f(x_i), w - x_i \rangle + f(x_i) + \Psi(w)] + V_{x_\omega}(w), \\ \sum_{i=1}^t \lambda_i^t &= \sum_{i=1}^t \frac{a_i}{A_t} = 1. \end{aligned} \quad (5.5.19)$$

3⁰. We need the following simple

Lemma 5.5.1 *Let $h : X \rightarrow \mathbf{R} \cup \{+\infty\}$ be a lower semicontinuous convex function which is finite on the relative interior of X , and let*

$$\bar{w} \in \operatorname{argmin}_{x \in X}^\epsilon \{h(x) + \omega(x)\}.$$

Then $\bar{w} \in X^o$, and for all $w \in X$ we have

$$h(w) + \omega(w) \geq h(\bar{w}) + \omega(\bar{w}) + V_{\bar{w}}(w) - \epsilon. \quad (5.5.20)$$

Proof. Invoking the definition of $\operatorname{argmin}^\epsilon$, we get $\bar{w} \in X^o$, and for some $h'(\bar{w}) \in \partial_X h(\bar{w})$ we have

$$\langle h'(\bar{w}) + \omega'(\bar{w}), w - \bar{w} \rangle \geq -\epsilon \quad \forall w \in X,$$

implying by convexity that

$$\begin{aligned} \forall w \in X : \\ h(w) + \omega(w) &\geq [h(\bar{w}) + \langle h'(\bar{w}), w - \bar{w} \rangle] + \omega(w) \\ &= [h(\bar{w}) + \omega(\bar{w}) + \underbrace{\langle h'(\bar{w}) + \omega'(\bar{w}), w - \bar{w} \rangle}_{\geq -\epsilon}] + V_{\bar{w}}(w), \end{aligned}$$

as required. \square

4⁰. Denote $\psi_t^* = \psi_t(z_t)$. Let us prove by induction that

$$B_t + \psi_t^* \geq A_t \phi(y_t^+), \quad t \geq 0, \quad B_t = (\epsilon + \bar{\epsilon})t. \quad (*_t)$$

For $t = 0$ this inequality is valid, since by initialization rules $\psi_0(\cdot) \geq 0$, $B_0 = 0$, and $A_0 = 0$. Suppose that $(*_t)$ holds true, and let us prove that so is $(*_t+1)$.

Taking into account (5.5.8) and the structure of $\psi_t(\cdot)$ as presented in (5.5.16), Lemma 5.5.1 as applied with $\bar{w} = z_t$ and $h(x) = \psi_t(x) - \omega(x) = \ell_t(x) + \alpha_t \Psi(x)$ yields, for all $w \in X$, the first inequality in the following chain:

$$\begin{aligned} \psi_t(w) &\geq \psi_t^* + V_{z_t}(w) - \epsilon \\ &\geq A_t \phi(y_t^+) + V_{z_t}(w) - \epsilon - B_t \quad [\text{by } (*_t)] \\ &\geq A_t \left[f(x_{t+1}) + \langle \nabla f(x_{t+1}), y_t^+ - x_{t+1} \rangle + \Psi(y_t^+) \right] + V_{z_t}(w) - \epsilon - B_t \quad [\text{since } f \text{ is convex}]. \end{aligned}$$

Invoking (5.5.13.b), the resulting inequality implies that for all $w \in X$ it holds

$$\begin{aligned} \psi_{t+1}(w) &\geq V_{z_t}(w) + A_t \left[f(x_{t+1}) + \langle \nabla f(x_{t+1}), y_t^+ - x_{t+1} \rangle + \Psi(y_t^+) \right] \\ &\quad + a_{t+1} [f(x_{t+1}) + \langle \nabla f(x_{t+1}), w - x_{t+1} \rangle + \Psi(w)] - \epsilon - B_t. \end{aligned} \quad (5.5.21)$$

By (5.5.11) and (5.5.10) we have $A_t(y_t^+ - x_{t+1}) - a_{t+1}x_{t+1} = -a_{t+1}z_t$, whence (5.5.21) implies that

$$\psi_{t+1}(w) \geq V_{z_t}(w) + A_{t+1}f(x_{t+1}) + A_t\Psi(y_t^+) + a_{t+1}[\langle \nabla f(x_{t+1}), w - z_t \rangle + \Psi(w)] - \epsilon - B_t \quad (5.5.22)$$

for all $w \in X$. Thus,

$$\begin{aligned} \psi_{t+1}^* &= \min_{w \in X} \psi_{t+1}(w) \\ &\stackrel{=h(w)+\omega(w) \text{ with convex lower semicontinuous } h}{\geq} \min_{w \in X} \{ \overbrace{V_{z_t}(w) + A_{t+1}f(x_{t+1}) + A_t\Psi(y_t^+) + a_{t+1}[\langle \nabla f(x_{t+1}), w - z_t \rangle + \Psi(w)]}^{} \} - \epsilon - B_t \\ &\quad [\text{by (5.5.22)}] \\ &\geq V_{z_t}(\hat{x}_{t+1}) + A_{t+1}f(x_{t+1}) + A_t\Psi(y_t^+) + a_{t+1}[\langle \nabla f(x_{t+1}), \hat{x}_{t+1} - z_t \rangle + \Psi(\hat{x}_{t+1})] \\ &\quad \underbrace{- \epsilon - \bar{\epsilon} - B_t}_{-B_{t+1}} \\ &\quad [\text{by (5.5.12) and Lemma 5.5.1}] \\ &= V_{z_t}(\hat{x}_{t+1}) + \underbrace{[A_t + a_{t+1}]}_{=A_{t+1} \text{ by (5.5.10)}} \left[\underbrace{\frac{A_t}{A_t + a_{t+1}} \Psi(y_t^+) + \frac{a_{t+1}}{A_t + a_{t+1}} \Psi(\hat{x}_{t+1})}_{\geq \Psi(y_{t+1}) \text{ by (5.5.13.a) and (5.5.10); recall that } \Psi \text{ is convex}} \right. \\ &\quad \left. + A_{t+1}f(x_{t+1}) + a_{t+1}\langle \nabla f(x_{t+1}), \hat{x}_{t+1} - z_t \rangle - B_{t+1} \right] \\ &\geq V_{z_t}(\hat{x}_{t+1}) + A_{t+1}f(x_{t+1}) + a_{t+1}\langle \nabla f(x_{t+1}), \hat{x}_{t+1} - z_t \rangle + A_{t+1}\Psi(y_{t+1}) - B_{t+1} \end{aligned}$$

that is,

$$\psi_{t+1}^* \geq V_{z_t}(\hat{x}_{t+1}) + A_{t+1}f(x_{t+1}) + a_{t+1}\langle \nabla f(x_{t+1}), \hat{x}_{t+1} - z_t \rangle + A_{t+1}\Psi(y_{t+1}) - B_{t+1}. \quad (5.5.23)$$

Now note that $\hat{x}_{t+1} - z_t = (y_{t+1} - x_{t+1})/\tau_t$ by (5.5.11) and (5.5.13.a), whence (5.5.23) implies

the first inequality in the following chain:

$$\begin{aligned}
\psi_{t+1}^* &\geq V_{z_t}(\hat{x}_{t+1}) + A_{t+1}f(x_{t+1}) + \frac{a_{t+1}}{\tau_t} \langle \nabla f(x_{t+1}), y_{t+1} - x_{t+1} \rangle + A_{t+1}\Psi(y_{t+1}) - B_{t+1} \\
&= V_{z_t}(\hat{x}_{t+1}) + A_{t+1}f(x_{t+1}) + A_{t+1} \langle \nabla f(x_{t+1}), y_{t+1} - x_{t+1} \rangle + A_{t+1}\Psi(y_{t+1}) - B_{t+1} \\
&\quad [\text{by (5.5.10)}] \\
&= A_{t+1} \left[\frac{1}{A_{t+1}} V_{z_t}(\hat{x}_{t+1}) + f(x_{t+1}) + \langle \nabla f(x_{t+1}), y_{t+1} - x_{t+1} \rangle + \Psi(y_{t+1}) \right] - B_{t+1} \\
&= A_{t+1}[\phi(y_{t+1}) + \delta_t] - B_{t+1} \quad [\text{by (5.5.14)}] \\
&\geq A_{t+1}\phi(y_{t+1}) - B_{t+1} \quad [\text{since we are in the case of } \delta_t \geq 0] \\
&\geq A_{t+1}\phi(y_{t+1}^+) - B_{t+1} \quad [\text{since } \phi(y_{t+1}^+) \leq \phi(y_{t+1})].
\end{aligned}$$

The inductive step is completed.

5⁰. Observe that $V_{z_t}(\hat{x}_{t+1}) \geq \frac{1}{2}\|\hat{x}_{t+1} - z_t\|^2$ and that $\hat{x}_{t+1} - z_t = (y_{t+1} - x_{t+1})/\tau_t$, implying that $V_{z_t}(\hat{x}_{t+1}) \geq \frac{1}{2\tau_t^2}\|y_{t+1} - x_{t+1}\|^2$, whence

$$\begin{aligned}
\delta_t &= \left[\frac{1}{A_{t+1}} V_{z_t}(\hat{x}_{t+1}) + \langle \nabla f(x_{t+1}), y_{t+1} - x_{t+1} \rangle + f(x_{t+1}) \right] - f(y_{t+1}) \quad [\text{by (5.5.14)}] \\
&\geq \left[\frac{1}{2A_{t+1}\tau_t^2} \|y_{t+1} - x_{t+1}\|^2 + \langle \nabla f(x_{t+1}), y_{t+1} - x_{t+1} \rangle + f(x_{t+1}) \right] - f(y_{t+1}) \\
&= \frac{L_t}{2} \|y_{t+1} - x_{t+1}\|^2 + \langle \nabla f(x_{t+1}), y_{t+1} - x_{t+1} \rangle + f(x_{t+1}) - f(y_{t+1}) \quad [\text{by (5.5.18)}]
\end{aligned}$$

whence $\delta_t \geq 0$ whenever $L_t = L_f$ by (5.5.3), as claimed.

6⁰. We have proved inequality $(*_t)$ for all $t \geq 0$. On the other hand,

$$\psi_t^* = \min_{w \in X} \psi_t(w) \leq \psi_t(x_*) \leq A_t\phi_* + V_{x_\omega}(x_*),$$

where the concluding inequality is due to (5.5.19). The resulting inequality taken together with $(*_t)$ yields

$$\phi(y_t^+) \leq \phi_* + A_t^{-1}[V_{x_\omega}(x_*) + B_t], \quad t = 1, 2, \dots \quad (5.5.24)$$

Now, from $A_0 = 0$, (5.5.9) and (5.5.10) it immediately follows that if $L_t \leq L$ for some L and all t and if \bar{A}_t are given by the recurrence

$$\bar{A}_0 = 0; \quad \bar{A}_{t+1} = \bar{A}_t + \left[\frac{1}{2L} + \sqrt{\frac{1}{4L^2} + \frac{\bar{A}_t}{L}} \right], \quad (5.5.25)$$

then $A_t \geq \bar{A}_t$ for all t . It is immediately seen that (5.5.25) implies that $\bar{A}_t \geq \frac{t^2}{4L}$ for all t ¹⁴. Taking into account that $L_t \leq L_f$ for all t by Theorem's premise, the bottom line is that $A_t \geq \frac{t^2}{4L_f}$ for all t , which combines with (5.5.24) to imply (5.5.15). \square

5.5.2 “Universal” Fast Gradient Methods

A conceptual shortcoming of the Fast Composite Gradient Method FGM presented in section 5.5.1 is that it is “tuned” to composite minimization problems (5.5.1) with smooth (with Lipschitz continuous gradient) functions f . We also know what to do when f is convex and just Lipschitz continuous; here we can use Mirror Descent. Note that Lipschitz continuity of the

¹⁴The simplest way to see the inequality is to set $2L\bar{A}_t = C_t^2$ and to note that (5.5.25) implies that $C_0 = 0$ and $C_{t+1}^2 = C_t^2 + 1 + \sqrt{1 + 2C_t^2} > C_t^2 + 2\sqrt{1/2}C_t + 1/2 = (C_t + \sqrt{1/2})^2$, so that $C_t \geq t/\sqrt{2}$ and thus $A_t \geq \frac{t^2}{4L}$ for all t .

gradient of a convex function f and Lipschitz continuity of f itself are “two extremes” in characterizing function’s smoothness; they can be linked together by “in-between” requirements for the gradient f' of f to be Hölder continuous with a given Hölder exponent $\kappa \in [1, 2]$:

$$\|f'(x) - f'(y)\|_* \leq L_\kappa \|x - y\|^{\kappa-1}, \quad \forall x, y \in X. \quad (5.5.26)$$

Inspired by Yu. Nesterov’s recent paper [32], we are about to modify the algorithm from section 5.5.1 to become applicable to problems of composite minimization (5.5.1) with functions f of smoothness varying between the outlined two extremes. It should be stressed that the resulting algorithm does *not* require a priori knowledge of smoothness parameters of f and adjusts itself to the actual level of smoothness as characterized by Hölder scale.

While inspired by [32], the algorithm to be developed is not completely identical to the one proposed in this reference; in particular, we still are able to work with inexact prox mappings.

5.5.2.1 Problem formulation

We continue to consider the optimization problem

$$\phi_* = \min_{x \in X} \{\phi(x) := f(x) + \Psi(x)\}, \quad (5.5.1)$$

where

1. X is a closed convex set in a Euclidean space E equipped with norm $\|\cdot\|$ (not necessarily the Euclidean one). Same as in section 5.5.1, we assume that X is equipped with distance-generating function $\omega(x)$ compatible with $\|\cdot\|$, and ,
2. $\Psi(x) : X \rightarrow \mathbf{R} \cup \{+\infty\}$ is a lower semicontinuous convex function which is finite on the relative interior of X ,
3. $f : X \rightarrow \mathbf{R}$ is a convex Lipschitz function satisfying, for some $\kappa \in [1, 2]$, $L \in (0, \infty)$, and a selection $f'(\cdot) : X \rightarrow E$ of subgradients, the relation

$$\forall (x, y \in X) : f(y) \leq f(x) + \langle f'(x), y - x \rangle + \frac{L}{\kappa} \|y - x\|^\kappa. \quad (5.5.27)$$

Observe that in section 5.5.1 we dealt with the case $\kappa = 2$ of the outlined situation. Note also that a *sufficient* (*necessary and sufficient* when $\kappa = 2$) condition for (5.5.27) is Hölder continuity of the gradient of f , i.e., the relation (5.5.26).

It should be stressed that *the algorithm to be built does not require a priori knowledge of κ and L* .

From now on we assume that (5.5.1) is solvable and denote by x_* an optimal solution to this problem.

In the sequel, we continuos to use our standard “prox-related” notation x_ω , X^o , $V_x(y)$, same as the notation

$$\operatorname{argmin}_{x \in X}^{\epsilon} \{h(x) + \omega(x)\} = \{z \in X^o : \exists h'(z) \in \partial_X h(z) : \langle h'(z) + \omega'(z), w - z \rangle \geq -\epsilon \quad \forall w \in X\}$$

see p. 384.

The algorithm we are about to develop requires the ability to find, given $\xi \in E$ and $\alpha \geq 0$, a point from the set

$$\operatorname{argmin}_{x \in X}^{\nu} \{\langle \xi, x \rangle + \alpha \Psi(x) + \omega(x)\}$$

where $\nu \geq 0$ is construction’s parameter.

5.5.2.2 Algorithm and Main Result

The *Fast Universal Composite Gradient Method*, nicknamed FUGM, we are about to consider is aimed at solving (5.5.1) within a given accuracy $\epsilon > 0$; we will refer to ϵ as to *target accuracy* of FUGM. On the top of its target accuracy ϵ , FUGM is specified by three other parameters

$$\nu \geq 0, \bar{\nu} \geq 0, \underline{L} > 0.$$

The method works in *stages*. The stages differ from each other only in their numbers of steps; for stage s , $s = 1, 2, \dots$, this number is $N_s = 2^s$.

Single Stage. N -step stage of FUGM is as follows.

- **Initialization:** Set $y_0 = x_\omega := \operatorname{argmin}_{x \in X} \omega(x)$, $\psi_0(w) = V_{x_\omega}(w)$, and select $y_0^+ \in X$ such that $\phi(y_0^+) \leq \phi(y_0)$. Set $A_0 = 0$.
- **Outer step** $t = 0, 1, \dots, N - 1$. Given $\psi_t(\cdot)$, $y_t^+ \in X$ and $A_t \geq 0$,

1. Compute

$$z_t \in \operatorname{argmin}_{w \in X}^\nu \psi_t(w). \quad (5.5.28)$$

2. **Inner Loop:** Set $L_{t,0} = \underline{L}$. For $\mu = 0, 1, \dots$, carry out the following computations.

- (a) Set $L_{t,\mu} = 2^\mu L_{t,0}$;
- (b) Find the positive root $a_{t+1,\mu}$ of the quadratic equation

$$L_{t,\mu} a_{t+1}^2 = A_t + a_{t+1,\mu} \quad \left[\Leftrightarrow a_{t+1,\mu} = \frac{1}{2L_{t,\mu}} + \sqrt{\frac{1}{4L_{t,\mu}^2} + \frac{A_t}{L_{t,\mu}}} \right] \quad (5.5.29)$$

and set

$$A_{t+1,\mu} = A_t + a_{t+1,\mu}, \quad \tau_{t,\mu} = \frac{a_{t+1,\mu}}{A_{t+1,\mu}}. \quad (5.5.30)$$

Note that $\tau_{t,\mu} \in (0, 1]$.

- (c) Set

$$x_{t+1,\mu} = \tau_{t,\mu} z_t + (1 - \tau_{t,\mu}) y_t^+. \quad (5.5.31)$$

and compute $f(x_{t+1,\mu})$, $f'(x_{t+1,\mu})$.

- (d) Compute

$$\hat{x}_{t+1,\mu} \in \operatorname{argmin}_{w \in X}^{\bar{\nu}} [a_{t+1,\mu} [\langle f'(x_{t+1,\mu}), w \rangle + \Psi(w)] + V_{z_t}(w)]. \quad (5.5.32)$$

- (e) Set

$$y_{t+1,\mu} = \tau_{t,\mu} \hat{x}_{t+1,\mu} + (1 - \tau_{t,\mu}) y_t^+, \quad (5.5.33)$$

compute $f(y_{t+1,\mu})$ and then compute

$$\begin{aligned} \delta_{t,\mu} &= \left[\frac{L_{t,\mu}}{2} \|y_{t+1,\mu} - x_{t+1,\mu}\|^2 + \langle f'(x_{t+1,\mu}), y_{t+1,\mu} - x_{t+1,\mu} \rangle + f(x_{t+1,\mu}) \right] \\ &\quad - f(y_{t+1,\mu}). \end{aligned} \quad (5.5.34)$$

(f) If $\delta_{t,\mu} < -\frac{\epsilon}{N}$, pass to the step $\mu + 1$ of Inner loop, otherwise terminate Inner loop and set

$$\begin{aligned} L_t &= L_{t,\mu}, & a_{t+1} &= a_{t+1,\mu}, & A_{t+1} &= A_{t+1,\mu}, & \tau_t &= \tau_{t,\mu}, \\ x_{t+1} &= x_{t+1,\mu}, & y_{t+1} &= y_{t+1,\mu}, & \delta_t &= \delta_{t,\mu}, \\ \psi_{t+1}(w) &= \psi_t(w) + a_{t+1} [f(x_{t+1}) + f'(x_{t+1}), w - x_{t+1}] + \Psi(w) \end{aligned} \quad (5.5.35)$$

3. Select somehow $y_{t+1}^+ \in X$ in such a way that $\phi(y_{t+1}^+) \leq \phi(y_{t+1})$.

Step t is completed. If $t < N - 1$, pass to outer step $t + 1$, otherwise terminate the stage and output y_N^+

Main result on the presented algorithm is as follows:

Theorem 5.5.2 Let function f in (5.5.1) satisfy (5.5.27) with some $L \in (0, \infty)$ and $\kappa \in [1, 2]$. As a result of an N -step stage \mathcal{S}_N of FUGM we get a point $y_N^+ \in X$ such that

$$\phi(y_N^+) - \phi_* \leq \epsilon + A_N^{-1} V_{x_\omega}(x_*) + [\nu + \bar{\nu}] N A_N^{-1}. \quad (5.5.36)$$

Furthermore,

$$\begin{aligned} (a) \quad \underline{L} &\leq L_t \leq \max[\underline{L}, 2L^{\frac{2}{\kappa}}[N/\epsilon]^{\frac{2-\kappa}{\kappa}}], \quad 0 \leq t \leq N-1, \\ (b) \quad A_N &\geq \left(\sum_{\tau=0}^{N-1} \frac{1}{2\sqrt{L_\tau}}\right)^2, \end{aligned} \quad (5.5.37)$$

Besides this, the number of steps of Inner loop at every outer step of \mathcal{S}_N does not exceed

$$M(N) := 1 + \log_2 \left(\max \left[1, 2\underline{L}^{-1} L^{\frac{2}{\kappa}} [N/\epsilon]^{\frac{2-\kappa}{\kappa}} \right] \right). \quad (5.5.38)$$

For proof, see the concluding subsection of this section.

Remark 5.5.1 Along with the outlined “aggressive” stepsize policy, where every Inner loop of a stage is initialized by setting $L_{t,0} = \underline{L}$, one can consider a “safe” stepsize policy where $L_{0,0} = \underline{L}$ and $L_{t,0} = L_{t-1}$ for $t > 0$. Inspecting the proof of Theorem 5.5.2 it is immediately seen that with the safe stepsize policy, (5.5.36) and (5.5.37) remain true, and the guaranteed bound on the total number of Inner loop steps at a stage (which is $N M(N)$ for the aggressive stepsize policy) becomes $N + M(N)$.

Assume from now on that

We have at our disposal a valid upper bound Θ_* on $V_{x_\omega}(x_*)$.

Corollary 5.5.1 Let function f in (5.5.1) satisfy (5.5.27) with some $L \in (0, \infty)$ and $\kappa \in [1, 2]$, and let

$$N_*(\epsilon) = \max \left[2 \frac{\sqrt{\underline{L}\Theta_*}}{\sqrt{\epsilon}}, \left(\frac{8^{\frac{\kappa}{2}} L \Theta_*^{\frac{\kappa}{2}}}{\epsilon} \right)^{\frac{2}{3\kappa-2}} \right]. \quad (5.5.39)$$

Whenever $N \geq N^*(\epsilon)$, the N -step stage \mathcal{S}_N of the algorithm ensures that

$$A_N^{-1} V_{x_\omega}(x_*) \leq A_N^{-1} \Theta_* \leq \epsilon. \quad (5.5.40)$$

In turn, the on-line verifiable condition

$$A_N^{-1} \Theta_* \leq \epsilon \quad (5.5.41)$$

implies, by (5.5.36), the relation

$$\phi(y_N^+) - \phi_* \leq 2\epsilon + [\nu + \bar{\nu}]NA_N^{-1}. \quad (5.5.42)$$

In particular, with exact prox-mappings (i.e., with $\nu = \bar{\nu} = 0$) it takes totally at most $\text{Ceil}(N^*(\epsilon))$ outer steps to ensure (5.5.41) and thus to ensure the error bound

$$\phi(y_N^+) - \phi_* \leq 2\epsilon.$$

Postponing for a moment Corollary's verification, let us discuss its consequences, for the time being – in the case $\nu = \bar{\nu} = 0$ of exact prox-mappings (the case of inexact mappings will be discussed in section 5.5.3).

Discussion. Corollary 5.5.1 suggests the following policy of solving problem (5.5.1) within a given accuracy $\epsilon > 0$: we run FUGM with somehow selected \underline{L} and $\nu = \bar{\nu} = 0$ stage by stage, starting with a single-step stage and increasing the number of steps N by factor 2 when passing from a stage to the next one. This process is continued until at the end of the current stage the condition , until at the end of the current stage the condition (5.5.41) (which is on-line verifiable) takes place. When it happens, we terminate; by Corollary 5.5.1, at this moment we have at our disposal a feasible 2ϵ -optimal solution to the problem of interest. On the other hand, the total, over all stages, number of Outer steps before termination clearly does not exceed twice the number of Outer steps at the concluding stage, that is, invoking Corollary 5.5.1, it does not exceed

$$N^*(\epsilon) := 4\text{Ceil}(N_*(\epsilon)).$$

As about the total, over all Outer steps of all stages, number M of Inner Loop steps (which is the actual complexity measure of FUGM), it, by Theorem 5.5.2, can be larger than $N^*(\epsilon)$ by at most by logarithmic in $\underline{L}, L, 1/\epsilon$ factor. Moreover, when “safe” stepsize policy is used, see Remark 5.5.1, we have by this Remark

$$M \leq \sum_{s \geq 0: 2^{s-1} \leq N_*(\epsilon)} [2^s + M(2^s)] \leq N^*(\epsilon) + \mathcal{M}, \quad \mathcal{M} := \sum_{s \geq 0: 2^{s-1} \leq N_*(\epsilon)} M(2^s)$$

with $M(N)$ given by (5.5.38). It is easily seen that for small ϵ the “logarithmic complexity term” \mathcal{M} is negligible as compared to the “leading complexity term” $N^*(\epsilon)$.

Further, setting $\bar{L} = \max[\underline{L}, L]$ and assuming $\epsilon \leq \bar{L}\Theta_*^{\kappa/2}$, we clearly have

$$N^*(\epsilon) \leq O(1) \left(\frac{\bar{L}\Theta_*^{\frac{\kappa}{2}}}{\epsilon} \right)^{\frac{2}{3\kappa-2}}. \quad (5.5.43)$$

Thus, our principal complexity term is expressed in terms of the true and unknown in advance smoothness parameters (L, κ) and is the better the better is smoothness (the larger is κ and the smaller is L). In the “most smooth case” $\kappa = 2$, we arrive at $N^*(\epsilon) = O(1)\sqrt{\frac{\bar{L}\Theta_*}{\epsilon}}$, which, basically, is the complexity bound for the Fast Composite Gradient Method FGM from section 5.5.1. In the “least smooth case” $\kappa = 1$, we get $N^*(\epsilon) = O(1)\frac{\bar{L}\sqrt{\Theta_*}}{\epsilon^2}$, which, basically, is the complexity bound of Mirror Descent¹⁵ Note that while in the nonsmooth case the complexity

¹⁵While we did not consider MD in the composite minimization setting, such a consideration is quite straightforward, and the complexity bounds for “composite MD” are identical to those of the “plain MD.”

bounds of FUGM and MD are “nearly the same,” the methods are essentially different, and in fact FUGM exhibits some theoretical advantage over MD: the complexity bound of FUGM depends on the maximal deviation L of the gradients of f from each other, the deviation being measured in $\|\cdot\|_*$, while what matters for MD is the Lipschitz constant, taken w.r.t. $\|\cdot\|_*$, of f , that is, the maximal deviation of the gradients of f from zero; the latter deviation can be much larger than the former one. Equivalently: the complexity bounds of MD are sensitive to adding to the objective a linear form, while the complexity bounds of FUGM are insensitive to such a perturbation.

Proof of Corollary 5.5.1. The first inequality in (5.5.40) is evident. Now let $N \geq N_*(\epsilon)$. Consider two possible cases, **I** and **II**, as defined below.

Case I: $2L^{\frac{2}{\kappa}}[N/\epsilon]^{\frac{2-\kappa}{\kappa}} \geq \underline{L}$. In this case, (5.5.37.a) ensures the first relation in the following chain:

$$\begin{aligned} L_t &\leq 2L^{\frac{2}{\kappa}}[N/\epsilon]^{\frac{2-\kappa}{\kappa}}, \quad 0 \leq t \leq N-1 \\ \Rightarrow A_N &\geq \left(2^{-3/2}L^{-\frac{1}{\kappa}}N^{1-\frac{2-\kappa}{2\kappa}}\epsilon^{\frac{2-\kappa}{2\kappa}}\right)^2 \quad [\text{by (5.5.37.b)}] \\ &= \frac{1}{8L^{\frac{2}{\kappa}}}N^{\frac{3\kappa-2}{\kappa}}\epsilon^{\frac{2-\kappa}{\kappa}} \\ &\geq \frac{1}{8L^{\frac{2}{\kappa}}}\left[\left(\frac{8^{\frac{\kappa}{2}}L\Theta_*^{\frac{\kappa}{2}}}{\epsilon}\right)^{\frac{2}{3\kappa-2}}\right]^{\frac{3\kappa-2}{\kappa}}\epsilon^{\frac{2-\kappa}{\kappa}} \quad [\text{by (5.5.39) and due to } N \geq N_*(\epsilon)] \\ &= \frac{1}{8L^{\frac{2}{\kappa}}}8L^{\frac{2}{\kappa}}\Theta_*\epsilon^{\frac{2-\kappa}{\kappa}-\frac{2}{\kappa}} = \Theta_*/\epsilon, \end{aligned}$$

and the second inequality in (5.5.40) follows.

Case II: $2L^{\frac{2}{\kappa}}[N/\epsilon]^{\frac{2-\kappa}{\kappa}} < \underline{L}$. In this case, (5.5.37.a) ensures the first relation in the following chain:

$$\begin{aligned} L_t &\leq \underline{L}, \quad 0 \leq t \leq N-1 \\ \Rightarrow A_N &\geq \frac{N^2}{4\underline{L}} \quad [\text{by (5.5.37.b)}] \\ &\geq \Theta_*/\epsilon \quad [\text{by (5.5.39) and due to } N \geq N_*(\epsilon)] \end{aligned}$$

and the second inequality in (5.5.40) follows. \square

5.5.2.3 Proof of Theorem 5.5.2

0⁰. By (5.5.27) we have

$$\delta_{t,\mu} \geq \frac{L_{t,\mu}}{2}\|y_{t+1,\mu} - x_{t+1,\mu}\|^2 - \frac{L}{\kappa}\|y_{t+1,\mu} - x_{t+1,\mu}\|^\kappa \geq \min_{r \geq 0} \left[\frac{L_{t,\mu}}{2}r^2 - \frac{L}{\kappa}r^\kappa \right]. \quad (5.5.44)$$

Assuming $\kappa = 2$, we see that $\delta_{t,\mu}$ is nonpositive whenever $L_{t,\mu} \geq L$, so that in the case in question Inner loop is finite, and

$$L_t \leq \max[\underline{L}, 2L]. \quad (5.5.45)$$

When $\kappa \in [1, 2)$, (5.5.44) after straightforward calculation implies that whenever

$$L_{t,\mu} \geq L^{\frac{2}{\kappa}}[N/\epsilon]^{\frac{2-\kappa}{\kappa}},$$

we have $\delta_{t,\mu} \geq -\frac{\epsilon}{N}$. Consequently, in the case in question Inner loop as finite as well, and

$$L_t \leq \max[\underline{L}, 2L^{\frac{2}{\kappa}}[N/\epsilon]^{\frac{2-\kappa}{\kappa}}], \quad (5.5.46)$$

Note that by (5.5.45) the resulting inequality holds true for $\kappa = 2$ as well.

The bottom line is that **FUGM** is well defined and maintains the following relations:

(a.1)	$\underline{L} \leq L_t \leq \max[\underline{L}, 2L^{\frac{2}{\kappa}}[N/\epsilon]^{\frac{2-\kappa}{\kappa}}]$,	
(a.2)	$a_{t+1} = \frac{1}{2L_t} + \sqrt{\frac{1}{4L_t^2} + \frac{A_t}{L_t}} \quad [\Leftrightarrow L_t a_{t+1}^2 = A_t + a_{t+1}],$ $A_0 = 0,$	
(a.3)	$A_{t+1} = \sum_{\tau=1}^{t+1} a_{\tau},$	
(a.4)	$\tau_t = \frac{a_{t+1}}{A_{t+1}} \in (0, 1],$	
(a.5)	$\frac{1}{\tau_t^2 A_{t+1}} = L_t \quad [\text{by (a.2-4)}];$	
(b.1)	$\psi_0(w) = V_{x_\omega}(w),$	
(b.2)	$\psi_{t+1}(w) = \psi_t(w) + a_{t+1} [f(x_{t+1}) + f'(x_{t+1}), w - x_{t+1}] + \Psi(w)]$	(5.5.47)
(c.1)	$z_t \in \underset{w \in X}{\operatorname{argmin}} \psi_t(w);$	
(c.2)	$x_{t+1} = \tau_t z_t + (1 - \tau_t) y_t^+,$	
(c.3)	$\hat{x}_{t+1} \in \underset{w \in X}{\operatorname{argmin}} [a_{t+1} [\langle f'(x_{t+1}), w \rangle + \Psi(w)] + V_{z_t}(w)],$	
(c.4)	$y_{t+1} = \tau_t \hat{x}_{t+1} + (1 - \tau_t) y_t^+;$	
(d)	$\frac{L_t}{2} \ y_{t+1} - x_{t+1}\ ^2 + \langle f'(x_{t+1}), y_{t+1} - x_{t+1} \rangle + f(x_{t+1}) - f(y_{t+1}) \geq -\frac{\epsilon}{N}$ [by termination rule for Inner loop]	
(e)	$y_t^+ \in X \ \& \ \phi(y_t^+) \leq \phi(y_t).$	

and, on the top of it, the number M of steps of Inner loop does not exceed

$$M(N) := 1 + \log_2 \left(\max \left[1, 2\underline{L}^{-1} L^{\frac{2}{\kappa}} [N/\epsilon]^{\frac{2-\kappa}{\kappa}} \right] \right). \quad (5.5.48)$$

1⁰. Observe that by (5.5.47.b) we have

$$\psi_t(\cdot) = \omega(\cdot) + \ell_t(\cdot) + \alpha_t \Psi(\cdot) \quad (5.5.49)$$

with nonnegative α_t and affine $\ell_t(\cdot)$, whence, taking into account that $\tau_t \in (0, 1]$, the algorithm maintains the inclusions $z_t \in X^o$, $x_t, \hat{x}_{t+1}, y_t, y_t^+ \in X$.

2⁰. Observe that by (5.5.47.a.2-4) we have

$$A_t = \sum_{i=1}^t a_i \quad (5.5.50)$$

(as always, the empty sum is 0) and

$$\frac{1}{2\tau_t^2 A_{t+1}} = \frac{A_{t+1}}{2a_{t+1}^2} = \frac{A_t + a_{t+1}}{2a_{t+1}^2} = \frac{L_t}{2}, \quad t \geq 0. \quad (5.5.51)$$

Observe also that (5.5.47.b1-2) and (5.5.50) imply that whenever $t \geq 1$, setting $\lambda_i^t = a_i/A_t$ we have

$$\begin{aligned} \psi_t(w) &= \sum_{i=1}^t a_i [\langle f'(x_i), w - x_i \rangle + f(x_i) + \Psi(w)] + V_{x_\omega}(w) \\ &= A_t \sum_{i=1}^t \lambda_i^t [\langle f'(x_i), w - x_i \rangle + f(x_i) + \Psi(w)] + V_{x_\omega}(w), \\ \sum_{i=1}^t \lambda_i^t &= \sum_{i=1}^t \frac{a_i}{A_t} = 1. \end{aligned} \quad (5.5.52)$$

3⁰. Denote $\psi_t^* = \psi_t(z_t)$. Let us prove by induction that

$$B_t + \psi_t^* \geq A_t \phi(y_t^+), \quad t \geq 0, \quad B_t = (\nu + \bar{\nu})t + \sum_{\tau=1}^t A_\tau \frac{\epsilon}{N}. \quad (*_t)$$

For $t = 0$ this inequality is valid, since by initialization rules $\psi_0(\cdot) \geq 0$, $B_0 = 0$, and $A_0 = 0$. Suppose that $(*_t)$ holds true for some $t \geq 0$, and let us prove that so is $(*_t+1)$.

Taking into account (5.5.47.c.1) and the structure of $\psi_t(\cdot)$ as presented in (5.5.49), Lemma 5.5.1 as applied with $\bar{w} = z_t$ and $h(x) = \psi_t(x) - \omega(x) = \ell_t(x) + \alpha_t \Psi(x)$ yields, for all $w \in X$, the first inequality in the following chain:

$$\begin{aligned} \psi_t(w) &\geq \psi_t^* + V_{z_t}(w) - \nu \\ &\geq A_t \phi(y_t^+) + V_{z_t}(w) - \nu - B_t \quad [\text{by } (*_t)] \\ &\geq A_t \left[f(x_{t+1}) + \langle f'(x_{t+1}), y_t^+ - x_{t+1} \rangle + \Psi(y_t^+) \right] + V_{z_t}(w) - \nu - B_t \quad [\text{since } f \text{ is convex}]. \end{aligned}$$

Invoking (5.5.47.b.2), the resulting inequality implies that for all $w \in X$ it holds

$$\begin{aligned} \psi_{t+1}(w) &\geq V_{z_t}(w) + A_t \left[f(x_{t+1}) + \langle f'(x_{t+1}), y_t^+ - x_{t+1} \rangle + \Psi(y_t^+) \right] \\ &\quad + a_{t+1} [f(x_{t+1}) + \langle f'(x_{t+1}), w - x_{t+1} \rangle + \Psi(w)] - \nu - B_t. \end{aligned} \quad (5.5.53)$$

By (5.5.47.c.2) and (5.5.47.a.4-5) we have $A_t(y_t^+ - x_{t+1}) - a_{t+1}x_{t+1} = -a_{t+1}z_t$, whence (5.5.53) implies that

$$\psi_{t+1}(w) \geq V_{z_t}(w) + A_{t+1}f(x_{t+1}) + A_t\Psi(y_t^+) + a_{t+1}[\langle f'(x_{t+1}), w - z_t \rangle + \Psi(w)] - \nu - B_t \quad (5.5.54)$$

for all $w \in X$. Thus,

$$\begin{aligned} \psi_{t+1}^* &= \min_{w \in X} \psi_{t+1}(w) \\ &\stackrel{=h(w)+\omega(w) \text{ with convex lower semicontinuous } h}{\geq} \min_{w \in X} \{ V_{z_t}(w) + A_{t+1}f(x_{t+1}) + A_t\Psi(y_t^+) + a_{t+1}[\langle f'(x_{t+1}), w - z_t \rangle + \Psi(w)] \} - \nu - B_t \\ &\quad [\text{by (5.5.54)}] \\ &\geq V_{z_t}(\hat{x}_{t+1}) + A_{t+1}f(x_{t+1}) + A_t\Psi(y_t^+) + a_{t+1}[\langle f'(x_{t+1}), \hat{x}_{t+1} - z_t \rangle + \Psi(\hat{x}_{t+1})] \\ &\quad - \nu - \bar{\nu} - B_t \\ &\quad [\text{by (5.5.47.c.3) and Lemma 5.5.1}] \\ &= V_{z_t}(\hat{x}_{t+1}) + \underbrace{[A_t + a_{t+1}]}_{=A_{t+1} \text{ by (5.5.47.a.3)}} \left[\underbrace{\frac{A_t}{A_t + a_{t+1}} \Psi(y_t^+) + \frac{a_{t+1}}{A_t + a_{t+1}} \Psi(\hat{x}_{t+1})}_{\geq \Psi(y_{t+1}) \text{ by (5.5.47.c.4) and (5.5.47.a.2-3); recall that } \Psi \text{ is convex}} \right. \\ &\quad \left. + A_{t+1}f(x_{t+1}) + a_{t+1}\langle f'(x_{t+1}), \hat{x}_{t+1} - z_t \rangle - \nu - \bar{\nu} - B_t \right] \\ &\geq V_{z_t}(\hat{x}_{t+1}) + A_{t+1}f(x_{t+1}) + a_{t+1}\langle f'(x_{t+1}), \hat{x}_{t+1} - z_t \rangle + A_{t+1}\Psi(y_{t+1}) - \nu - \bar{\nu} - B_t \end{aligned}$$

that is,

$$\psi_{t+1}^* \geq V_{z_t}(\hat{x}_{t+1}) + A_{t+1}f(x_{t+1}) + a_{t+1}\langle f'(x_{t+1}), \hat{x}_{t+1} - z_t \rangle + A_{t+1}\Psi(y_{t+1}) - \nu - \bar{\nu} - B_t. \quad (5.5.55)$$

Now note that

$$\hat{x}_{t+1} - z_t = (y_{t+1} - x_{t+1})/\tau_t \quad (5.5.56)$$

by (5.5.47.c), whence (5.5.55) implies the first inequality in the following chain:

$$\begin{aligned}
\psi_{t+1}^* &\geq V_{z_t}(\hat{x}_{t+1}) + A_{t+1}f(x_{t+1}) + \frac{a_{t+1}}{\tau_t} \langle f'(x_{t+1}), y_{t+1} - x_{t+1} \rangle + A_{t+1}\Psi(y_{t+1}) - \nu - \bar{\nu} - B_t \\
&= V_{z_t}(\hat{x}_{t+1}) + A_{t+1}f(x_{t+1}) + A_{t+1} \langle f'(x_{t+1}), y_{t+1} - x_{t+1} \rangle + A_{t+1}\Psi(y_{t+1}) - \nu - \bar{\nu} - B_t \\
&\quad [\text{by (5.5.47.a.4)}] \\
&= A_{t+1} \left[\frac{1}{A_{t+1}} V_{z_t}(\hat{x}_{t+1}) + f(x_{t+1}) + \langle f'(x_{t+1}), y_{t+1} - x_{t+1} \rangle + \Psi(y_{t+1}) \right] - \nu - \bar{\nu} - B_t \\
&= A_{t+1} \left[\phi(y_{t+1}) + \left[\frac{1}{A_{t+1}} V_{z_t}(\hat{x}_{t+1}) + f(x_{t+1}) + \langle f'(x_{t+1}, y_{t+1} - x_{t+1}) - f(y_{t+1}) \rangle \right] \right] \\
&\quad - \nu - \bar{\nu} - B_t \\
&\geq A_{t+1} \left[\phi(y_{t+1}) + \left[\frac{1}{2A_{t+1}} \|\hat{x}_{t+1} - z_t\|^2 + f(x_{t+1}) + \langle f'(x_{t+1}, y_{t+1} - x_{t+1}) - f(y_{t+1}) \rangle \right] \right] \\
&\quad - \nu - \bar{\nu} - B_t \\
&= A_{t+1}\phi(y_{t+1}) \\
&\quad + A_{t+1} \left[\frac{1}{2\tau_t^2 A_{t+1}} \|x_{t+1} - y_{t+1}\|^2 + f(x_{t+1}) + \langle f'(x_{t+1}, y_{t+1} - x_{t+1}) - f(y_{t+1}) \rangle \right] \\
&\quad - \nu - \bar{\nu} - B_t \quad [\text{by (5.5.56)}] \\
&= A_{t+1} \left[\phi(y_{t+1}) + \left[\frac{L_t}{2} \|x_{t+1} - y_{t+1}\|^2 + f(x_{t+1}) + \langle f'(x_{t+1}, y_{t+1} - x_{t+1}) - f(y_{t+1}) \rangle \right] \right] \\
&\quad - \nu - \bar{\nu} - B_t \quad [\text{by (5.5.47.a.5)}] \\
&\geq A_{t+1}\phi(y_{t+1}) - A_{t+1}\frac{\epsilon}{N} - \nu - \bar{\nu} - B_t \quad [\text{by (5.5.47.d)}] \\
&= A_{t+1}\phi(y_{t+1}) - B_{t+1} \quad [\text{see } (*_t)] \\
&\geq A_{t+1}\phi(y_{t+1}^+) - B_{t+1} \quad [\text{by (5.5.47.e)}].
\end{aligned}$$

The inductive step is completed.

4⁰. We have proved inequality $(*_t)$ for all $t \geq 0$. On the other hand,

$$\psi_t^* = \min_{w \in X} \psi_t(w) \leq \psi_t(x_*) \leq A_t \phi_* + V_{x_\omega}(x_*),$$

where the concluding inequality is due to (5.5.52). The resulting inequality taken together with $(*_t)$ yields

$$\begin{aligned}
\phi(y_t^+) - \phi_* &\leq A_t^{-1}[V_{x_\omega}(x_*) + B_t] \\
&= A_t^{-1}V_{x_\omega}(x_*) + A_t^{-1} \sum_{\tau=1}^t A_\tau \frac{\epsilon}{N} + A_t^{-1}[\nu + \bar{\nu}]t \\
&\leq A_t^{-1}V_{x_\omega}(x_*) + \epsilon \frac{t}{N} + A_t^{-1}[\nu + \bar{\nu}]t, \quad t = 1, 2, \dots, N,
\end{aligned} \tag{5.5.57}$$

where the concluding \leq is due to $A_\tau \leq A_t$ for $\tau \leq t$.

5⁰. From $A_0 = 0$ and (5.5.47.a.2) by induction in t it follows that

$$A_t \geq \underbrace{\left(\sum_{\tau=0}^{t-1} \frac{1}{2\sqrt{L_\tau}} \right)^2}_{=: \bar{A}_t}. \tag{5.5.58}$$

Indeed, (5.5.58) clearly holds true for $t = 0$. Assuming that the relation holds true for some t and invoking (5.5.47.a.2), we get

$$\begin{aligned}
A_{t+1} &= A_t + a_{t+1} \geq \bar{A}_t + \sqrt{\bar{A}_t/L_t} + \frac{1}{2L_t} \geq \left[\sum_{\tau=0}^{t-1} \frac{1}{\sqrt{L_\tau}} \right]^2 + 2 \left[\sum_{\tau=0}^{t-1} \frac{1}{\sqrt{L_\tau}} \right] \frac{1}{2\sqrt{L_t}} + \frac{1}{4L_t} \\
&= \left[\sum_{\tau=0}^t \frac{1}{2\sqrt{L_\tau}} \right]^2 = \bar{A}_{t+1}^2,
\end{aligned}$$

completing the induction.

Bottom line. Combining (5.5.47.a.1), (5.5.57), (5.5.58) and taking into account (5.5.48), we arrive at Theorem 5.5.2. \square

5.5.3 From Fast Gradient Minimization to Conditional Gradient

5.5.3.1 Proximal and Linear Minimization Oracle based First Order algorithms

First Order algorithms considered so far heavily utilize proximal setups; to be implementable, these algorithms need the convex compact domain X of convex problem

$$\text{Opt} = \min_{x \in X} f(x) \quad (\text{CP})$$

to be *proximal friendly* – to admit a strongly convex (and thus nonlinear!) DGF $\omega(\cdot)$ with easy to minimize over X linear perturbations $\omega(X) + \langle \xi, x \rangle$ ¹⁶. Whenever this is the case, it is easy to minimize over X linear forms $\langle \xi, x \rangle$ (indeed, minimizing such a form over X within a whatever high accuracy reduces to minimizing over X the sum $\omega(x) + \alpha \langle \xi, x \rangle$ with large α). The inverse conclusion is, in general, not true: it may happen that X admits a computationally cheap *Linear Minimization Oracle* (LMO) – a procedure capable to minimize over X linear forms, while no DGFs leading to equally computationally cheap proximal mappings are known. The important for applications examples include:

1. nuclear norm ball $X \subset \mathbf{R}^{n \times n}$ playing crucial role in low rank matrix recovery: with all known proximal setups, computing prox-mappings requires computing full singular value decomposition of an $n \times n$ matrix. At the same time, minimizing a linear form over X requires finding just the pair of leading singular vectors of an $n \times n$ matrix x (why?) The latter task can be easily solved by Power method¹⁷ For n in the range of few thousands and more, computing the leading singular vectors is by order of magnitude cheaper, progressively as n grows, than the full singular value decomposition.
2. spectahedron $X = \{x \in \mathbf{S}^n : x \succeq 0, \text{Tr}(x) = 1\}$ “responsible” for Semidefinite Programming. Situation here is similar to the one with nuclear norm ball: with all known proximal setups, computing a proximal mapping requires full eigenvalue decomposition of a symmetric $n \times n$ matrix, while minimizing a linear form over X reduces to approximating the leading eigenvector of such a matrix; the latter task can be achieved by Power method and is, progressively in n , much cheaper computationally than full eigenvalue decomposition.
3. *Total Variation ball* X in the space of $n \times n$ images ($n \times n$ arrays with zero mean). Total Variation of an $n \times n$ image x is

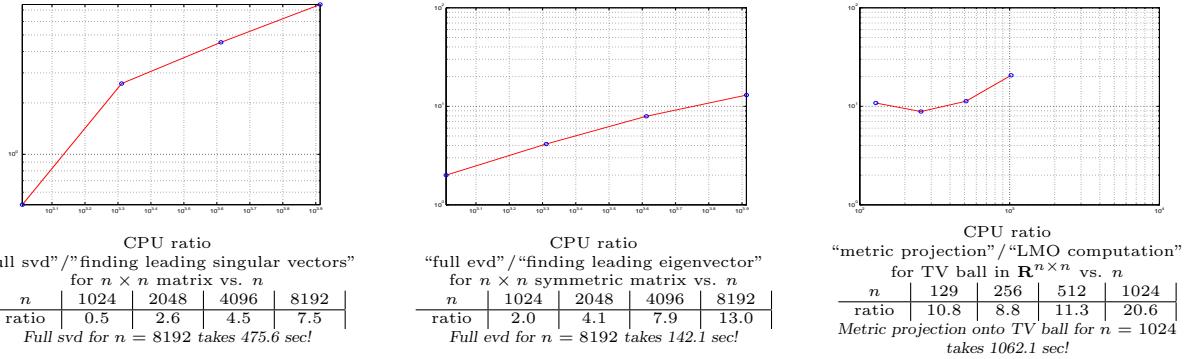
$$\text{TV}(x) = \sum_{i=1}^{n-1} \sum_{j=1}^n |x_{i+1,j} - x_{i,j}| + \sum_{i=1}^n \sum_{j=1}^{n-1} |x_{i,j+1} - x_{i,j}|$$

¹⁶On the top of proximal friendliness, we would like to ensure a moderate, “nearly dimension-independent” value of the ω -radius of X , in order to avoid rapid growth of the iteration count with problem’s dimension. Note, however, that in the absence of proximal friendliness, we cannot even start running a proximal algorithm...

¹⁷In the most primitive implementation, Power method iterates matrix-vector multiplications $e_t \mapsto e_{t+1} = x^T x e_t$, with randomly selected, usually from standard Gaussian distribution, starting vector e_0 ; after few tens of iterations, vectors e_t and $x e_t$ become good approximations of the left and the right leading singular vectors of x .

and is a norm on the space of images (the discrete analogy of L_1 norm of the gradient of a function on a 2D square); convex minimization over TV balls and related problems (like LASSO problem (5.5.4) with the space of images in the role of E and $\text{TV}(\cdot)$ in the role of $\|\cdot\|_E$) plays important role in image reconstruction. Computing the simplest – Euclidean – prox mapping on TV ball in the space of $n \times n$ images requires computing Euclidean projection of a point onto large-scale polytope cut off $O(1)n^2 \ell_1$ ball by $O(1)n^2$ homogeneous linear equality constraints; for n of practical interest (few hundreds), this problem is quite time consuming. In contrast, it turns out that minimizing a linear form over TV ball requires solving a special Maximum Flow problem [15], and this turns out to be by order of magnitudes cheaper than metric projection onto TV ball.

Here is some numerical data allowing to get an impression of the “magnitude” of the just outlined phenomena:



Platform: 2 × 3.40 GHz CPU, 16.0 GB RAM, 64-bit Windows 7

The outlined and some other generic large-scale problems of convex minimization inspire investigating the abilities to solve problems (CP) on LMO - represented domains X – convex compact domains equipped by Linear Minimization Oracles. As about the objective f , we still assume that it is given by First Order Oracle.

5.5.3.2 Conditional Gradient algorithm

The only classical optimization algorithm capable to handle convex minimization over LMO-represented domain is *Conditional Gradient Algorithm* (CGA) invented as early as in 1956 by Frank and Wolfe [11]. CGA does not work when the function f in (CP) possesses no smoothness beyond Lipschitz continuity, so in the sequel we impose some “nontrivial smoothness” assumption. More specifically, we consider the same composite minimization problem

$$\phi_* = \min_{x \in X} \{\phi(x) := f(x) + \Psi(x)\}, \quad (5.5.1)$$

as in the previous sections, and assume that

- X is a convex and compact subset in Euclidean space E equipped with a norm $\|\cdot\|$,
- Ψ is convex lower semicontinuous on X and is finite on the relative interior of X function, and
- f is convex and continuously differentiable on X function satisfying, for some $L < \infty$ and $\kappa \in (1, 2]$ (note that $\kappa > 1$!) relation (5.5.27):

$$\forall (x, y \in X) : f(y) \leq f(x) + \langle f'(x), y - x \rangle + \frac{L}{\kappa} \|y - x\|^\kappa.$$

Composite LMO and Composite Conditional Gradient algorithm. We assume that we have at our disposal a *Composite Linear Minimization Oracle* (CLMO) – a procedure which, given on input a linear form $\langle \xi, x \rangle$ on the embedding X Euclidean space E and a nonnegative α , returns a point

$$\text{CLMO}(\xi, \alpha) \in \operatorname{Argmin}_{x \in X} \{\langle \xi, x \rangle + \alpha \Psi(x)\}. \quad (5.5.59)$$

The associated Composite Conditional Gradient Algorithm (nicknamed in the sequel CGA) as applied to (5.5.1) generates iterates $x_t \in X$, $t = 1, 2, \dots$, as follows:

1. *Initialization:* x_1 is an arbitrary points of X .

2. *Step $t = 1, 2, \dots$:* Given x_t , we

- compute $f'(x_t)$ and call CLMO to get the point

$$x_t^+ = \text{CLMO}(f'(x_t), 1) \quad [\in \operatorname{Argmin}_{x \in X} [\ell_t(x) := f(x_t) + \langle f'(x_t), x - x_t \rangle + \Psi(x)]] \quad (5.5.60)$$

- compute the point

$$\hat{x}_{t+1} = x_t + \gamma_t(x_t^+ - x_t), \quad \gamma_t = \frac{2}{t+1} \quad (5.5.61)$$

- take as x_{t+1} a (whatever) point in X such that $\phi(x_{t+1}) \leq \phi(\hat{x}_{t+1})$.

The convergence properties of the algorithm are summarized in the following simple statement going back to [11]:

Theorem 5.5.3 *Let Composite CGA be applied to problem (5.5.1) with convex f satisfying (5.5.27) with some $\kappa \in (1, 2]$, and let D be the $\|\cdot\|$ -diameter of X . Denoting by ϕ_* the optimal value in (5.5.1), for all $t \geq 2$ one has*

$$\epsilon_t := \phi(x_t) - \phi_* \leq \frac{2LD^\kappa}{\kappa(3-\kappa)} \gamma_t^{\kappa-1} = \frac{2\kappa LD^\kappa}{\kappa(3-\kappa)(t+1)^{\kappa-1}}. \quad (5.5.62)$$

Proof. Let

$$\ell_t(x) = f(x_t) + \langle f'(x_t), x - x_t \rangle + \Psi(x).$$

We have

$$\begin{aligned} \epsilon_{t+1} &:= \phi(x_{t+1}) - \phi_* \leq \phi(\hat{x}_{t+1}) - \phi_* = \phi(x_t + \gamma_t(x_t^+ - x_t)) - \phi_* \\ &= \ell_t(x_t + \gamma_t(x_t^+ - x_t)) + [f(x_t + \gamma_t(x_t^+ - x_t)) - f(x_t) - \langle f'(x_t), \gamma_t(x_t^+ - x_t) \rangle] - \phi_* \\ &\leq [(1 - \gamma_t) \underbrace{\ell(x_t)}_{=\phi(x_t)} + \gamma_t \ell(x_t^+) - \phi_*] + \frac{L}{\kappa} [\gamma_t D]^\kappa \\ &\quad [\text{since } \ell_t(\cdot) \text{ is convex and due to (5.5.27); note that } \|\gamma_t(x_t^+ - x_t)\| \leq \gamma_t D] \\ &= [(1 - \gamma_t) \underbrace{[\phi(x_t) - \phi_*]}_{\epsilon_t} + \gamma_t [\ell(x_t^+) - \phi_*]] + \frac{L}{\kappa} [\gamma_t D]^\kappa \\ &= (1 - \gamma_t) \epsilon_t + \frac{L}{\kappa} [\gamma_t D]^\kappa + \gamma_t [\ell(x_t^+) - \phi_*] \\ &\leq (1 - \gamma_t) \epsilon_t + \frac{L}{\kappa} [\gamma_t D]^\kappa \\ &\quad [\text{since } \ell_t(x) \leq \phi(x) \text{ for all } x \in X \text{ and } \ell(x_t^+) = \min_{x \in X} \ell_t(x)] \end{aligned}$$

We have arrived at the recurrent inequality for the residuals $\epsilon_t = \phi(x_t) - \phi_*$:

$$\epsilon_{t+1} \leq (1 - \gamma_t) \epsilon_t + \frac{L}{\kappa} [\gamma_t D]^\kappa, \quad \gamma_t = \frac{2}{t+1}, \quad t = 1, 2, \dots \quad (5.5.63)$$

Let us prove by induction in $t \geq 2$ that for $t = 2, 3, \dots$ it holds

$$\epsilon_t \leq \frac{2LD^\kappa}{\kappa(3-\kappa)} \gamma_t^{\kappa-1} \quad (*_t)$$

The base $t = 2$ is evident: from (5.5.63) it follows that $\epsilon_2 \leq \frac{LD^\kappa}{\kappa}$, and the latter quantity is $\leq \frac{2LD^\kappa}{\kappa(3-\kappa)} \gamma_2^{\kappa-1} = \frac{2^\kappa LD^\kappa}{\kappa(3-\kappa)}$ ¹⁸. Assuming that $(*_t)$ holds true for some $t \geq 2$, we get from (5.5.63) the first inequality in the following chain:

$$\begin{aligned} \epsilon_{t+1} &\leq \frac{2LD^\kappa}{\kappa(3-\kappa)} [\gamma_t^{\kappa-1} - \gamma_t^\kappa] + \frac{LD^\kappa}{\kappa} \gamma_t^\kappa \\ &= \frac{2LD^\kappa}{(3-\kappa)\kappa} \left[\gamma_t^{\kappa-1} - \left[1 - \frac{3-\kappa}{2} \right] \gamma_t^\kappa \right] = \frac{2LD^\kappa}{(3-\kappa)\kappa} 2^{\kappa-1} \left[(t+1)^{-(\kappa-1)} + (1-\kappa)(t+1)^{-\kappa} \right] \\ &\leq \frac{2LD^\kappa}{(3-\kappa)\kappa} 2^{\kappa-1} (t+2)^{-\kappa-1}, \end{aligned}$$

where the concluding inequality in the chain is the gradient inequality for the convex function $u^{1-\kappa}$ of $u > 0$. We see that the validity of $(*_t)$ for some t implies the validity of $(*_t)$; induction, and thus the proof of Theorem 5.5.3, is complete. \square

5.5.3.3 Bridging Fast and Coniditonal Gradient algorithms

Now assume that we are interested in problem (5.5.1) in the situation considered in section 5.5.2, so that X is a convex subset in Euclidean space E with a norm $\|\cdot\|$, and X is equipped with a compatible with this norm DGF ω . From now on, we make two additional assumptions:

- A. X is a compact set, $\omega(\cdot)$ is continuously differentiable on X (so that $X^o = X$), and for some known $\Upsilon < \infty$ one has

$$\forall (z \in X, w \in X) : |\langle \omega'(z), w - z \rangle| \leq \Upsilon. \quad (5.5.64)$$

Remark: For all standard proximal setups described in section 5.3.3, except for the Entropy one, one can take as Υ the quantity $O(1)\Theta$, with Θ defined in section 5.3.2.

- B. We have at our disposal a Composite Linear Minimization Oracle (5.5.59).

Lemma 5.5.2 *Under assumptions A, B, let*

$$\epsilon = \Upsilon, \bar{\epsilon} = 3\Upsilon.$$

Then for every $\xi \in E$, $\alpha \geq 0$ and every $z \in X$ one has

- (a) $\text{CLMO}(\xi, \alpha) \in \underset{x \in X}{\operatorname{argmin}} \epsilon \{ \langle \xi, x \rangle + \alpha \Psi(x) + \omega(x) \},$
- (b) $\text{CLMO}(\xi, \alpha) \in \underset{x \in X}{\operatorname{argmin}} \bar{\epsilon} \{ \langle \xi, x \rangle + \alpha \Psi(x) + V_z(x) \}.$

Proof. Given $\xi \in E$, $\alpha > 0$, $z \in X$, and setting $w = \text{CLMO}(\xi, \alpha)$, we have $w \in X^o = X$ and for some $\Psi'(w) \in \partial_X \Psi(w)$ one has

$$\langle \xi + \alpha \Psi'(w), x - w \rangle \geq 0 \quad \forall x \in X.$$

¹⁸Indeed, the required inequality boils down to the inequality $3^\kappa - \kappa 3^{\kappa-1} \leq 2^\kappa$, which indeed is true when $\kappa \geq 1$, due to the convexity of the function u^κ on the ray $u \geq 0$.

Consequently,

$$\forall x \in X : \langle \xi + \alpha \Psi'(w) + \omega'(w), x - w \rangle \geq \langle \omega'(w), x - w \rangle \geq -\Upsilon,$$

whence $w \in \operatorname{argmin}_{x \in X}^{\epsilon} \{ \langle \xi, x \rangle + \alpha \Psi(x) + \omega(x) \}$. Similarly,

$$\begin{aligned} \forall x \in X : & \langle \xi + \alpha \Psi'(w) + \omega'(w) - \omega'(z), x - w \rangle \geq \langle \omega'(w) - \omega'(z), x - w \rangle \\ & \geq -\Upsilon + \langle \omega'(z), w - x \rangle = -\Upsilon + \langle \omega'(z), w - z \rangle + \langle \omega'(z), z - x \rangle \geq -3\Upsilon. \end{aligned}$$

whence $w \in \operatorname{argmin}_{x \in X}^{\bar{\epsilon}} \{ \langle \xi, x \rangle + \alpha \Psi(x) + V_z(w) \}$. The case of $\alpha = 0$ is completely similar (redefine Ψ as identically zero and α as 1). \square

Combining Lemma 5.5.2 and Theorem 5.5.1, we arrive at the following

Corollary 5.5.2 *In the situation described in section 5.5.1.1 and under assumptions **A**, **B**, consider the implementation of the Fast Composite Gradient Method (5.5.8) – (5.5.14) where (5.5.8) is implemented as*

$$z_t = \text{CLMO}(\nabla \ell_t(\cdot), \alpha_t) \quad (5.5.65)$$

(see (5.5.16)), and (5.5.12) is implemented as

$$\hat{x}_{t+1} = \text{CLMO}(a_{t+1} \nabla f(x_{t+1}), a_{t+1}) \quad (5.5.66)$$

(which, by Lemma 5.5.2, ensures (5.5.8), (5.5.12) with $\epsilon = \Upsilon$, $\bar{\epsilon} = 3\Upsilon$). Assuming that (5.5.7) holds true and that $\delta_t \geq 0$ for all t (the latter definitely is the case when $L_t = L_f$ for all t), one has

$$y_t, y_t^+ \in X, \phi(y_t^+) - \phi_* \leq \phi(y_t) - \phi_* \leq A_t^{-1} [V_{x_\omega}(x_*) + 4\Upsilon t] \leq \frac{4L_f}{t^2} [V_{x_\omega}(x_*) + 4\Upsilon t] \quad (5.5.67)$$

for $t = 1, 2, \dots$, where, as always, x_* is an optimal solution to (5.5.1) and x_ω is the minimizer of $\omega(\cdot)$ on X .

A step of the algorithm, modulo updates $y_t \mapsto y_t^+$, reduces to computing the value and the gradient of f at a point, computing the value of f at another point and to computing two values of Composite Linear Minimization Oracle, with no computations of the values and the derivatives of $\omega(\cdot)$ at all.

Discussion. Observe that (5.5.64) clearly implies that $V_{x_\omega}(x_*) \leq \Upsilon$, which combines with (5.5.67) to yield the relation

$$\phi(y_t^+) - \phi_* \leq O(1) \frac{L_f \Upsilon}{t}, \quad t = 1, 2, \dots \quad (5.5.68)$$

On the other hand, under the premise of Corollary 5.5.2, we have at our disposal a Composite LMO for X and the function f in (5.5.1) is convex and satisfies (5.5.27) with $L = L_f$ and $\kappa = 2$, see (5.5.2). As a result, we can solve the problem of interest (5.5.1) by CGA ; the resulting efficiency estimate, as stated by Theorem 5.5.3 in the case of $L = L_f$, $\kappa = 2$, is

$$\phi(x_t) - \phi_* \leq O(1) \frac{L_f D^2}{t}, \quad (5.5.69)$$

where D is the $\|\cdot\|$ -diameter of X . Assuming that

$$\Upsilon \leq \alpha(X)D^2 \quad (5.5.70)$$

for properly selected $\alpha(X) \geq 1$, the efficiency estimate (5.5.68) is by at most factor $O(1)\alpha(X)$ worse than the CGA efficiency estimate (5.5.69), so that for moderate $\alpha(X)$ the estimates are “nearly the same.” This is not too surprising, since FGM with imprecise proximal mappings given by (5.5.65), (5.5.66) is, basically, LMO-based rather than proximal algorithm. Indeed, with on-line tuning of L_t ’s, the implementation of FGM described in Corollary 5.5.2 requires operating with $\omega(z_t)$, $\omega'(z_t)$, and $\omega(\hat{x}_{t+1})$ solely in order to compute δ_t when checking whether this quantity is nonnegative. We could avoid the necessity to work with ω by computing $\|\hat{x}_{t+1} - z_t\|$ in order to build the lower bound

$$\bar{\delta}_t = \left[\frac{1}{2A_{t+1}} \|\hat{x}_{t+1} - z_t\|^2 + \langle \nabla f(x_{t+1}), y_{t+1} - x_{t+1} \rangle + f(x_{t+1}) \right] - f(y_{t+1})$$

on δ_t , and to use this bound in the role of δ_t when updating L_t ’s (note that from item 5⁰ of the proof of Theorem 5.5.1 it follows that $\bar{\delta}_t \geq 0$ whenever $L_t = L_f$). Finally, when both the norm $\|\cdot\|$ and $\omega(\cdot)$ are difficult to compute, we can use the off-line policy $L_t = L_f$ for all t , thus avoiding completely the necessity to check whether $\delta_t \geq 0$ and ending up with an algorithm, let it be called FGM-LMO, based solely on Composite Linear Minimization Oracle and never “touching” the DGF ω (which is used solely in the algorithm’s analysis). The efficiency estimate of the resulting algorithm is within $O(1)\alpha(X)$ factor of the one of CGA.

Note that in many important situations (5.5.70) indeed is satisfied with “quite moderate” $\alpha(X)$. For example, for the Euclidean setup one can take $\alpha(X) = 1$. For other standard proximal setups presented in section 5.3.3, except for the Entropy one, $\alpha(X)$ just logarithmically grows with the dimension of X .

An immediate question is, what, if any, do we gain when passing from simple and transparent CGA to much more sophisticated FGM-LMO. The answer is: as far as theoretical efficiency estimates are concerned, nothing is gained. Nevertheless, “bridging” proximal Fast Composite Gradient Method and LMO-based Conditional Gradients possesses some “methodological value.” For example, FGM-LMO replaces the precise composite prox mapping

$$(\xi, z) \mapsto \operatorname{argmin}_{x \in X} [\langle \xi, x \rangle + \alpha \Psi(x) + V_z(x)]$$

with its “most crude” approximation

$$(\xi, z) \mapsto \text{CLMO}(\xi, \alpha) \in \operatorname{Argmin}_{x \in X} [\langle \xi, x \rangle + \alpha \Psi(x)].$$

which ignores Bregman distance $V_z(x)$ at all. In principle, one could utilize a less crude approximation of the composite prox mapping, “shifting” the CGA efficiency estimate $O(LD^2/t)$ towards the $O(LD^2/t^2)$ efficiency estimate of FGM with precise proximal mappings – the option we here just mention rather than exploit in depth.

5.5.3.4 LMO-based implementation of Fast Universal Gradient Method

The above “bridging” of FGM and CGA as applied to Composite Minimization problem (5.5.1) with convex and “fully smooth” f (one with Lipschitz continuous gradient) can be extended to the case of less smooth f ’s, with FUGM in the role of FGM. Specifically, with assumptions **A**,

B in force, let us look at Composite Minimization problem (5.5.1) with convex f satisfying the smoothness requirement (5.5.27) with some $L > 0$ and $\kappa \in (1, 2]$ (pay attention to $\kappa > 1!$).

By Lemma 5.5.2, setting

$$\nu = \Upsilon, \bar{\nu} = 3\Upsilon.$$

we ensure that for every $\xi \in E$, $\alpha \geq 0$ and every $z \in X$ one has

$$\begin{aligned} (a) \quad \text{CLMO}(\xi, \alpha) &\in \operatorname{argmin}_{x \in X}^{\nu} \{ \langle \xi, x \rangle + \alpha \Psi(x) + \omega(x) \}, \\ (b) \quad \text{CLMO}(\xi, \alpha) &\in \operatorname{argmin}_{x \in X}^{\bar{\nu}} \{ \langle \xi, x \rangle + \alpha \Psi(x) + V_z(x) \}. \end{aligned} \quad (5.5.71)$$

Therefore we can consider the implementation of FUGM (in the sequel referred to as FUGM-LMO) where the target accuracy is a given ϵ , the parameters ν and $\bar{\nu}$ are specified as $\nu = \Upsilon$, $\bar{\nu} = 3\Upsilon$, and

- z_t is defined according to

$$z_t = \text{CLMO}(\nabla \ell_t(\cdot), \alpha_t).$$

with $\ell_t(\cdot)$, α_t defined in (5.5.49);

- $\hat{x}_{t+1,\mu}$ is defined according to

$$\hat{x}_{t+1,\mu} = \text{CLMO}(a_{t+1,\mu} f'(x_{t+1,\mu}), a_{t+1,\mu}).$$

By (5.5.71), this implementation ensures (5.5.28), (5.5.32).

Let us upper-bound the quantity $[\nu + \bar{\nu}]N A_N^{-1}$ appearing in (5.5.36) and (5.5.42).

Lemma 5.5.3 *Let assumption **A** take place, and let FUGM-LMO be applied to Composite Minimization problem (5.5.1) with convex function f satisfying the smoothness condition (5.5.27) with some $L \in (0, \infty)$ and $\kappa \in (1, 2]$. When the number of steps N at a stage of the method satisfies the relation*

$$N \geq N^\#(\epsilon) := \max \left[\frac{16L\Upsilon}{\epsilon}, 2^{\frac{5\kappa}{2(\kappa-1)}} \left(\frac{L\Upsilon^{\frac{\kappa}{2}}}{\epsilon} \right)^{\frac{1}{\kappa-1}} \right] \quad (5.5.72)$$

one has

$$[\nu + \bar{\nu}]N A_N^{-1} \leq \epsilon.$$

Proof. We have $A_N^{-1}N[\nu + \bar{\nu}] = 4\Upsilon A_N^{-1}N$. Assuming $N \geq N^\#(\epsilon)$, consider the same two cases as in the proof of Corollary 5.5.1. Similarly to this proof,

— in Case I we have

$$\begin{aligned} A_N &\geq \frac{1}{8L^{\frac{2}{\kappa}}} N^{\frac{3\kappa-2}{\kappa}} \epsilon^{\frac{\kappa-2}{\kappa}} \\ \Rightarrow 4A_N^{-1}N\Upsilon &\leq 32\Upsilon L^{\frac{2}{\kappa}} N^{\frac{2(1-\kappa)}{\kappa}} \epsilon^{\frac{\kappa-2}{\kappa}} \leq \epsilon \quad [\text{due to } N \geq 2^{\frac{5\kappa}{2(\kappa-1)}} \left(\frac{L\Upsilon^{\frac{\kappa}{2}}}{\epsilon} \right)^{\frac{1}{\kappa-1}}]; \end{aligned}$$

— in Case II we have

$$\begin{aligned} A_N &\geq \frac{N^2}{4L} \\ \Rightarrow 4A_N^{-1}N\Upsilon &\leq 16 \frac{L\Upsilon}{N} \leq \epsilon \quad [\text{due to } N \geq \frac{16L\Upsilon}{\epsilon}]. \end{aligned} \quad \square$$

Combining Corollary 5.5.1, Lemma 5.5.3 and taking into account that the quantity Υ participating in assumption **A** can be taken as Θ_* participating in Corollary 5.5.1 (see Discussion after Corollary 5.5.2), we arrive at the following result:

Corollary 5.5.3 Let assumption **A** take place, and let FUGM-LMO be applied to Composite Minimization problem (5.5.1) with convex function f satisfying, for some $L \in (0, \infty)$ and $\kappa \in (1, 2]$, the smoothness condition (5.5.27). The on-line verifiable condition

$$4A_N^{-1}N\Upsilon \leq \epsilon \quad (5.5.73)$$

ensure that the result y_N^+ of N -step stage \mathcal{S}_N of FUGM-LMO satisfies the relation

$$\phi(y_N^+) - \phi_* \leq 3\epsilon.$$

Relation (5.5.73) definitely is satisfied when

$$N \geq N(\epsilon) := \max[N_{\#}(\epsilon), N^{\#}(\epsilon)],$$

where

$$N_{\#}(\epsilon) = \max \left[2 \frac{\sqrt{L\Upsilon}}{\sqrt{\epsilon}}, \left(\frac{8^{\frac{\kappa}{2}} L \Upsilon^{\frac{\kappa}{2}}}{\epsilon} \right)^{\frac{2}{3\kappa-2}} \right] \quad (5.5.74)$$

(cf. (5.5.39)) and $N^{\#}(\epsilon)$ is given by (5.5.72).

Remarks. **I.** Note that FUGM-LMO does not require computing $\omega(\cdot)$ and $\omega'(\cdot)$ and operates solely with CLMO(\cdot, \cdot), First Order oracle for f , and the Zero Order oracle for $\|\cdot\|$ (the latter is used to compute $\delta_{t,\mu}$, see (5.5.34)). Assuming κ and L known in advance, we can further avoid the necessity to compute the norm $\|\cdot\|$ and to skip, essentially, the Inner loop. To this end it suffices to carry out a single step $\mu = 0$ of the Inner loop with $L_{t,0}$ set to $L^{\frac{2}{\kappa}}[N/\epsilon]^{\frac{2-\kappa}{\kappa}}$ and to skip computing $\delta_{t,0}$ – the latter quantity automatically is $\geq -\epsilon/N$, see item **1⁰** in the proof of Theorem 5.5.2, p. 396.

II. It is immediately seen that for small enough values of $\epsilon > 0$, we have $N^{\#}(\epsilon) \geq N_{\#}(\epsilon)$, so that the total, over all stages, number of outer steps before ϵ -solution is built is upper-bounded by $O(1)N^{\#}(\epsilon)$; with “safe” stepsize policy, see Remark 5.5.1, and small ϵ the total, over all stages, number of Inner steps also is upper-bounded by $O(1)N^{\#}(\epsilon)$. Note that in the case of (5.5.70) and for small enough $\epsilon > 0$, we have

$$N^{\#}(\epsilon) \leq [O(1)\alpha(X)]^{\frac{\kappa}{2(\kappa-1)}} \left(\frac{LD^{\kappa}}{\epsilon} \right)^{\frac{1}{\kappa-1}},$$

implying that the complexity bound of FUGM-LMO for small ϵ is at most by factor $[O(1)\alpha(X)]^{\frac{\kappa}{2(\kappa-1)}}$ worse than the complexity bound for CGA as given by (5.5.62). The conclusions one can make from these observations are completely similar to those related to FGM-LMO, see p. 404.

5.5.4 Saddle Point representations and Mirror Prox algorithm

5.5.4.1 Motivation

Let us revisit the reasons for our interest in the First Order methods. It stemmed from the fact that when solving large-scale convex programs, applying polynomial time algorithms becomes questionable: unless the matrices of the Newton-type linear systems to be solved at iterations of an IPM possess “favourable sparsity pattern,” a single IPM iteration in the large-scale case

becomes too time-consuming to be practical. Thus, we were looking for methods with cheap iterations, and stated that in the case of constrained nonsmooth problems (this is what we deal with in most of applications), the only “cheap” methods we know are First Order algorithms. We then observed that as a matter of fact, all these algorithms are black-box-oriented, and thus obey the limits of performance established by Information-based Complexity Theory, and focused on algorithms which achieve these limits of performance. A bad news here is that the “limits of performance” are quite poor; the best we can get here is the complexity like $N(\epsilon) = O(1/\epsilon^2)$. When one tries to understand why this indeed is the best we can get, it becomes clear that the reason is in “blindness” of the black-box-oriented algorithms we use; they utilize only local information on the problem (values and subgradients of the objective) and make no use of a priori knowledge of problem’s structure. Here it should be stressed that as far as Convex Programming is concerned, we nearly always do have detailed knowledge of problem’s structure – otherwise, how could we know that the problem is convex in the first place? The First Order methods as presented so far just do not know how to utilize problem’s structure, in sharp contrast with IPM’s which are fully “structure-oriented.” Indeed, this is the *structure* of an optimization problem which underlies its setting as an LP/CQP/SDP program, and IPM’s, same as the Simplex method, are quite far from operating with the values and subgradients of the objective and the constraints; instead, they operate directly on problem’s data. We can say that the LP/CQP/SDP framework is a “structure-revealing” one, and it allows for algorithms with fast convergence, algorithms capable to build a high accuracy solution in a moderate number of iterations. In contrast to this, the black-box-oriented framework is “structure-obscuring,” and the associated algorithms, al least in the large-scale case, possess really slow rate of convergence. The bad news here is that the LP/CQP/SDP framework requires iterations which in many applications are prohibitively time-consuming in the large scale case.

Now, for a long time the LP/CQP/SDP framework was the only one which allowed for utilizing problem’s structure. The situation changed significantly circa 2003, when Yu. Nesterov discovered a “computationally cheap” way to do so. The starting point in his discovery was the well known fact that the disastrous complexity $N(\epsilon) \geq O(1/\epsilon^2)$ in large-scale nonsmooth convex optimization comes from nonsmoothness; passing from nonsmooth problems (minimizing a convex Lipschitz continuous objective over a convex compact domain) to their smooth counterparts, where the objective, still black-box-represented one, has a Lipschitz continuous gradient, reduces the complexity bound to $O(1/\sqrt{\epsilon})$, and for smooth problems with favourable geometry, this reduced complexity turns out to be nearly- or fully dimension-independent, see section 5.5.1. Now, the “smooth” complexity $O(1/\sqrt{\epsilon})$, while still not a polynomial-time one, is a dramatic improvement as compared to its “nonsmooth” counterpart $O(1/\epsilon^2)$. All this was known for a couple of decades and was of nearly no practical use, since problems of minimizing a *smooth* convex objective over a *simple* convex set (the latter should be simple in order to allow for computationally cheap algorithms) are a “rare commodity”, almost newer met in applications. The major part of Nesterov’s 2003 breakthrough is the observation, surprisingly simple in the hindsight¹⁹, that

In convex optimization, the typical source of nonsmoothness is taking maxima, and

¹⁹Let me note that the number of actual breakthroughs which are “trivial in retrospect” is much larger than one could expect. For example, George Dantzig, the founder of Mathematical Programming, considered as one of his major contributions (and I believe, rightly so) the very formulation of an LP problem, specifically, the idea of adding to the constraints an objective to be optimized, instead of making decisions via “ground rules,” which was standard at the time.

usually a “heavily nonsmooth” convex objective $f(x)$ can be represented as

$$f(x) = \max_{y \in Y} \phi(x, y)$$

with a pretty simple (quite often, just bilinear) convex-concave function ϕ and simple convex compact set Y .

As an immediate example, look at the function $f(x) = \max_{1 \leq i \leq m} [a_i^T x + b_i]$ ²⁰; this function can be represented as $f(x) = \max_{y \in \Delta_m} \phi(x, y) = \sum_{i=1}^m y_i [a_i^T x + b_i]$, so that both ϕ and Y are as simple as they could be...

As a result, the nonsmooth problem

$$\min_{x \in X} f(x) \quad (\text{CP})$$

can be reformulated as the saddle point problem

$$\min_{x \in X} \max_{y \in Y} \phi(x, y) \quad (\text{SP})$$

with a pretty smooth (usually just bilinear) convex-concave cost function ϕ .

Now, saddle point reformulation (SP) of the problem of interest (CP) can be used in different ways. Nesterov himself used it to smoothen the function f . Specifically, he assumes (see [29]) that ϕ is of the form

$$\phi(x, y) = x^T (Ay + a) - \psi(y)$$

with convex ψ , and observes that then the function

$$f_\delta(x) = \max_{y \in Y} [\phi(x, y) - \delta d(y)],$$

where the “regularizer” $d(\cdot)$ is strongly convex, is smooth (possesses Lipschitz continuous gradient) and is close to f when δ is small. He then minimizes the smooth function f_δ by his optimal method for smooth convex minimization – one with the complexity $O(1/\sqrt{\epsilon})$. Since the Lipschitz constant of the gradient of f_δ deteriorates as $\delta \rightarrow +0$, on one hand, and we need to work with small δ , those of the order of accuracy ϵ to which we want to solve (CP), on the other hand, the overall complexity of the outlined *smoothing* method is $O(1/\epsilon)$, which still is much better than the “black-box nonsmooth” complexity $O(1/\epsilon^2)$ and in fact is, in many cases, as good as it could be.

For example, consider the Least Squares problem

$$\text{Opt} = \min_{x \in \mathbf{R}^n : \|x\|_2 \leq R} \|Bx - b\|_2, \quad (\text{LS})$$

where B is an $n \times n$ matrix of the spectral norm $\|B\|$ not exceeding L . This problem can be easily reformulated in the form of (SP), specifically, as

$$\min_{\|x\|_2 \leq 1} \max_{\|y\|_2 \leq 1} y^T [Bx - b],$$

²⁰these are exactly the functions responsible for the lower complexity bound $O(1/\epsilon^2)$.

and Nesterov's smoothing with $d(y) = \frac{1}{2}y^T y$ allows to find an ϵ -solution to the problem in $O(1)\frac{LR}{\epsilon}$ steps, provided that $LR/\epsilon \geq 1$; each step here requires two matrix-vector multiplications, one involving B and one involving B^T . On the other hand, it is known that when $n > LR/\epsilon$, every method for solving (LS) which is given in advance b and is allowed to "learn" B via matrix-vector multiplications involving B and B^T , needs in the worst (over b and B with $\|B\| \leq L$) case at least $O(1)LR/\epsilon$ matrix-vector multiplications in order to find an ϵ -solution to (LS). We can say that in the large-scale case complexity of the pretty well structured Least Squares problems (LS) with respect to First Order methods is $O(1/\epsilon)$, and this is what Nesterov's smoothing achieves.

Nesterov's smoothing is one way to utilize a "nice" (with "good" ϕ) saddle point representation (SP) of the objective in (CP) to accelerate processing the latter problem. In the sequel, we present another way to utilize (SP) – the *Mirror Prox* (MP) algorithm, which needs ϕ to be smooth (with Lipschitz continuous gradient), works directly on the saddle point problem (SP) and solves it with complexity $O(1/\epsilon)$, same as Nesterov's smoothing. Let us stress that the "fast gradient methods" (this seems to become a common name of the algorithms we have just mentioned and their numerous "relatives" developed in the recent years) do exploit problem's structure: this is exactly what is used to build the saddle point reformulation of the problem of interest; and since the reformulated problem is solved by black-box-oriented methods, we have good chances to end up with computationally cheap algorithms.

5.5.4.2 Examples of saddle point representations

Here we present instructive examples of "nice" – just bilinear – saddle point representations of important and "heavily nonsmooth" convex functions. Justification of the examples is left to the reader. Note that the list can be easily extended.

1. $\|\cdot\|_p$ -norm:

$$\|x\|_p = \max_{y: \|y\|_q \leq 1} \langle y, x \rangle, \quad q = \frac{p}{p-1}.$$

Matrix analogy: the Shatten p -norm of a rectangular matrix $x \in \mathbf{R}^{m \times n}$ is defined as $|x|_p = \|\sigma(x)\|_p$, where $\sigma(x) = [\sigma_1(x); \dots; \sigma_{\min[m,n]}(x)]$ is the vector of the largest $\min[m, n]$ singular values of x (i.e., the singular values which have chances to be nonzero) arranged in the non-ascending order. We have

$$\forall x \in \mathbf{R}^{m \times n} : |x|_p = \max_{y \in \mathbf{R}^{m \times n}, \|y\|_q \leq 1} \langle y, x \rangle \equiv \text{Tr}(y^T x), \quad q = \frac{p}{p-1}.$$

Variation: p -norm of the "positive part" $x_+ = [\max[x_1, 0]; \dots; \max[x_n, 0]]$ of a vector $x \in \mathbf{R}^n$:

$$\|x_+\|_p = \max_{y \in \mathbf{R}^n: y \geq 0, \|y\|_q \leq 1} \langle y, x \rangle \equiv y^T x, \quad q = \frac{p}{p-1}.$$

Matrix analogy: for a symmetric matrix $x \in \mathbf{S}^n$, let x_+ be the matrix with the same eigenvectors and eigenvalues replaced with their positive parts. Then

$$|x_+|_p = \max_{y \in \mathbf{S}^n: y \succeq 0, \|y\|_q \leq 1} \langle y, x \rangle = \text{Tr}(yx), \quad q = \frac{p}{p-1}.$$

2. Maximum entry $s_1(x)$ in a vector x and the sum $s_k(x)$ of k maximal entries in a vector $x \in \mathbf{R}^n$:

$$\begin{aligned}\forall(x \in \mathbf{R}^n, k \leq n) : \\ s_1(x) &= \max_{y \in \Delta_n} \langle y, x \rangle, \\ \Delta_n &= \{y \in \mathbf{R}^n : y \geq 0, \sum_i y_i = 1\}; \\ s_k(x) &= \max_{y \in \Delta_{n,k}} \langle y, x \rangle, \\ \Delta_{n,k} &= \{y \in \mathbf{R}^n : 0 \leq y_i \leq 1 \forall i, \sum_i y_i = k\}\end{aligned}$$

Matrix analogies: maximal eigenvalue $S_1(x)$ and the sum of k largest eigenvalues $S_k(x)$ of a matrix $x \in \mathbf{S}^n$:

$$\begin{aligned}\forall(x \in \mathbf{S}^n, k \leq n) : \\ S_1(x) &= \max_{y \in \mathcal{D}_n} \langle y, x \rangle \equiv \text{Tr}(yx), \\ \mathcal{D}_n &= \{y \in \mathbf{S}^n : y \succeq 0, \text{Tr}(y) = 1\}; \\ S_k(x) &= \max_{y \in \mathcal{D}_{n,k}} \langle y, x \rangle \equiv \text{Tr}(yx), \\ \mathcal{D}_{n,k} &= \{y \in \mathbf{S}^n : 0 \preceq y \preceq I, \text{Tr}(y) = k\}\end{aligned}$$

3. Maximal magnitude $s_1(\text{abs}[x])$ of entries in x and the sum $s_k(\text{abs}[x])$ of the k largest magnitudes of entries in x :

$$\begin{aligned}\forall(x \in \mathbf{R}^n, k \leq n) : \\ s_1(\text{abs}[x]) &= \max_{y \in \mathbf{R}^n, \|y\|_1 \leq 1} \langle y, x \rangle; \\ s_k(\text{abs}[x]) &= \max_{y \in \mathbf{R}^n : \|y\|_\infty \leq 1, \|y\|_1 \leq k} \langle y, x \rangle \equiv y^T x.\end{aligned}$$

Matrix analogy: the sum $\Sigma_k(x)$ of k maximal singular values of a matrix x :

$$\forall(x \in \mathbf{R}^{m \times n}, k \leq \min[m, n]) : \Sigma_k(x) = \max_{y \in \mathbf{R}^{m \times n} : \|y\|_\infty \leq 1, \|y\|_1 \leq k} \langle y, x \rangle \equiv \text{Tr}(y^T x).$$

We augment this sample of “raw materials” with calculus rules. We restrict ourselves with *bilinear saddle point representations*:

$$f(x) = \max_{y \in Y} [\langle y, Ax \rangle + \langle a, x \rangle + \langle b, y \rangle + c],$$

where x runs through a Euclidean space E_x , Y is a convex compact subset in a Euclidean space E_y , $x \mapsto Ax : E_x \rightarrow E_y$ is a linear mapping, and $\langle \cdot, \cdot \rangle$ are inner products in the respective spaces. In the rules to follow, all Y_i are nonempty convex *compact* sets.

1. [Taking conic combinations] Let

$$f_i(x) = \max_{y_i \in Y_i} [\langle y_i, A_i x \rangle + \langle a_i, x \rangle + \langle b_i, y_i \rangle + c_i], \quad 1 \leq i \leq m,$$

and let $\lambda_i \geq 0$. Then

$$\begin{aligned}&\sum_i \lambda_i f_i(x) \\ &= \max_{y=[y_1; \dots; y_m] \in Y=Y_1 \times \dots \times Y_m} [\underbrace{\sum_i \lambda_i \langle y_i, A_i x \rangle}_{=: \langle y, Ax \rangle} + \underbrace{\sum_i \lambda_i a_i, x \rangle}_{=: \langle a, x \rangle} + \underbrace{\sum_i \lambda_i \langle b_i, y_i \rangle}_{=: \langle b, y \rangle} + \underbrace{\sum_i \lambda_i c_i}_{=: c}].\end{aligned}$$

2. [Taking direct sums] Let

$$f_i(x_i) = \max_{y_i \in Y_i} [\langle y_i, A_i x_i \rangle + \langle a_i, x_i \rangle + \langle b_i, y_i \rangle + c_i], \quad 1 \leq i \leq m.$$

Then

$$\begin{aligned} f(x \equiv [x^1; \dots; x^m]) &= \sum_{i=1}^m f_i(x_i) \\ &= \max_{y=[y_1; \dots; y_m] \in Y=Y_1 \times \dots \times Y_m} \left[\underbrace{\sum_i \lambda_i \langle y_i, A_i x_i \rangle}_{=: \langle y, Ax \rangle} + \underbrace{\sum_i \lambda_i \langle a_i, x_i \rangle}_{=: \langle a, x \rangle} + \underbrace{\sum_i \lambda_i \langle b_i, y_i \rangle}_{=: \langle b, y \rangle} + \underbrace{\sum_i \lambda_i c_i}_{=: c} \right]. \end{aligned}$$

3. [Affine substitution of argument] For $x \in \mathbf{R}^n$, let

$$f(x) = \max_{y \in Y} [\langle y, Ax \rangle + \langle a, x \rangle + \langle b, y \rangle + c],$$

and let $\xi \mapsto P\xi + p$ be an affine mapping from \mathbf{R}^m to \mathbf{R}^n . Then

$$g(\xi) := f(P\xi + p) = \max_{y \in Y} [\langle y, AP\xi \rangle + \langle P^*a, \xi \rangle + \langle b, y \rangle + [c + \langle a, p \rangle]].$$

4. [Taking maximum] Let

$$f_i(x) = \max_{y_i \in Y_i} [\langle y_i, A_i x \rangle + \langle a_i, x \rangle + \langle b_i, y_i \rangle + c_i], \quad 1 \leq i \leq m.$$

Then

$$\begin{aligned} f(x) &:= \max_i f_i(x) \\ &= \max_{y=[(\eta_1, u_1); \dots; (\eta_m, u_m)] \in Y} \left[\underbrace{\left\{ \sum_i \langle u_i, A_i x \rangle + \langle \sum_i \eta_i a_i, x \rangle \right\}}_{=: \langle y, Ax \rangle} + \underbrace{\left\{ \sum_i \langle b_i, u_i \rangle + \sum_i c_i \eta_i \right\}}_{=: \langle b, y \rangle} \right], \\ Y &= \left\{ y = [(\eta_1, u_1); \dots; (\eta_m, u_m)] : (\eta_i, u_i) \in \widehat{Y}_i, 1 \leq i \leq m, \sum_i \eta_i = 1 \right\}, \end{aligned}$$

where \widehat{Y}_i are the “conic hats” spanned by Y_i :

$$\widehat{Y}_i = \text{cl} \left\{ (\eta, u) : 0 < \eta \leq 1, \eta^{-1}u \in Y_i \right\} = \{[0, 0]\} \cup \left\{ (\eta, u) : 0 < \eta \leq 1, \eta^{-1}u \in Y_i \right\}$$

(the second “=” follows from the compactness of Y_i). Here is the derivation:

$$\begin{aligned} \max_i f_i(x) &= \max_{\eta \in \Delta_m} \sum_i \eta_i f_i(x) \\ &= \max_{\eta \in \Delta_m, y_i \in Y_i, 1 \leq i \leq m} \left[\sum_i \langle \eta_i y_i, A_i x \rangle + \sum_i \langle \eta_i a_i, x \rangle + \sum_i \langle b_i, \eta_i y_i \rangle + \sum_i \eta_i c_i \right] \\ &= \max_{(\eta_i, u_i) \in \widehat{Y}_i, 1 \leq i \leq m, \sum_i \eta_i = 1} \left[\left[\sum_i \langle u_i, A_i x \rangle + \langle \sum_i \eta_i a_i, x \rangle \right] + \left[\sum_i \langle b_i, u_i \rangle + \sum_i c_i \eta_i \right] \right]. \end{aligned}$$

5.5.4.3 The Mirror Prox algorithm

We are about to present the *Mirror Prox algorithm* [21] for solving saddle point problem

$$\min_{x \in X} \max_{y \in Y} \phi(x, y) \tag{SP}$$

with smooth convex-concave cost function ϕ . Smoothness means that ϕ is continuously differentiable on $Z = X \times Y$, and that the associated vector field (cf. (5.3.47))

$$F(z = (x, y)) = [F_x(x, y) := \nabla_x \phi(x, y); F_y(x, y) = -\nabla_y \phi(x, y)] : Z := X \times Y \rightarrow \mathbf{R}^{n_x} \times \mathbf{R}^{n_y}$$

is Lipschitz continuous on Z .

The setup for the MP algorithm as applied to (SP) is the same as for the saddle point Mirror Descent, see section 5.3.6, specifically, we fix a norm $\|\cdot\|$ on the embedding space $\mathbf{R}^{n_x} \times \mathbf{R}^{n_y}$ of the domain Z of ϕ and a DGF ω for this domain. Since F is Lipschitz continuous, we have for properly chosen $L < \infty$:

$$\|F(z) - F(z')\|_* \leq L\|z - z'\| \quad \forall z, z' \in Z. \quad (5.5.75)$$

The MP algorithm as applied to (SP) is the recurrence

$$\begin{aligned} z_1 := (x_1, y_1) &= z_\omega := \operatorname{argmin}_{z \in Z} \omega(z), \\ w_t := (\tilde{x}_t, \tilde{y}_t) &= \operatorname{Prox}_{z_t}(\gamma_t F(z_t)) \\ z_{t+1} := (x_{t+1}, y_{t+1}) &= \operatorname{Prox}_{z_t}(\gamma_t F(w_t)), \\ &\quad \operatorname{Prox}_z(\xi) = \operatorname{argmin}_{w \in Z} [V_z(w) + \langle \xi, w \rangle], \\ &\quad V_z(w) = \omega(w) - \langle \omega'(z), w - z \rangle - \omega(z) \geq \frac{1}{2}\|w - z\|^2, \\ z^t &= \left[\sum_{\tau=1}^t \gamma_\tau \right]^{-1} \sum_{\tau=1}^t \gamma_\tau w_\tau, \end{aligned} \quad (5.5.76)$$

where γ_t are positive stepsizes. This looks pretty much as the saddle point MD algorithm, see (5.3.48); the only difference is in what is called *extra-gradient* step: what was the updating $z_t \mapsto z_{t+1}$ in the MD, now defines an intermediate point w_t , and the actual updating $z_t \mapsto z_{t+1}$ uses, instead of the value of F at z_t , the value of F at this intermediate point. Another difference is that now the approximate solutions z^t are weighted sums of the intermediate points w_τ , and not of the subsequent prox-centers z_τ , as in MD. These “minor changes” result in major consequences (provided that F is Lipschitz continuous). Here is the corresponding result:

Theorem 5.5.4 *Let convex-concave saddle point problem (SP) be solved by the MP algorithm. Then*

(i) *For every τ , one has for all $u \in Z$:*

$$\begin{aligned} \gamma_\tau \langle F(w_\tau), w_\tau - u \rangle &\leq V_{z_\tau}(u) - V_{z_{\tau+1}}(u) + \underbrace{[\gamma_\tau \langle F(w_\tau), w_\tau - z_{\tau+1} \rangle - V_{z_\tau}(z_{\tau+1})]}_{=: \delta_\tau}, \quad (a) \\ \delta_\tau &\leq \gamma_\tau \langle F(z_{\tau+1}) - F(w_\tau), w_\tau - z_{\tau+1} \rangle - V_{z_\tau}(w_\tau) - V_{w_\tau}(z_{\tau+1}) \quad (b) \\ &\leq \frac{\gamma_\tau^2}{2} \|F(w_\tau) - F(z_{\tau+1})\|_*^2 - \frac{1}{2} \|w_\tau - z_{\tau+1}\|^2 \quad (c) \end{aligned} \quad (5.5.77)$$

In particular, in the case of (5.5.75) one has

$$\gamma_\tau \leq \frac{1}{L} \Rightarrow \delta_\tau \leq 0,$$

see (5.5.77.c).

(ii) *For $t = 1, 2, \dots$, one has*

$$\epsilon_{\text{sad}}(z^t) \leq \frac{\frac{\Omega^2}{2} + \sum_{\tau=1}^t \delta_\tau}{\sum_{\tau=1}^t \gamma_\tau}. \quad (5.5.78)$$

In particular, let the stepsizes γ_τ , $1 \leq \tau \leq t$, be such that $\gamma_\tau \geq \frac{1}{L}$ and $\delta_\tau \leq 0$ for all $\tau \leq t$ (by (i), this is definitely so for the stepsizes $\gamma_\tau \equiv \frac{1}{L}$, provided that (5.5.75) takes place). Then

$$\epsilon_{\text{sad}}(z^t) \leq \frac{\Omega^2}{2 \sum_{\tau=1}^t \gamma_\tau} \leq \frac{\Omega^2 L}{2t}. \quad (5.5.79)$$

Proof. 1^0 . To verify (i), let us fix τ and set $\delta = \delta_\tau$, $z = z_\tau$, $\xi = \gamma_\tau F(z_\tau)$, $w = w_\tau$, $\eta = \gamma_\tau F(w_\tau)$, $z_+ = z_{\tau+1}$. Note that the relations between the entities in question are given by

$$z \in Z^o, w = \text{Prox}_z(\xi), z_+ = \text{Prox}_z(\eta). \quad (5.5.80)$$

As we remember, from $w = \text{Prox}_z(\xi)$ and $z_+ = \text{Prox}_z(\eta)$ it follows that $w, z_+ \in Z^o$ and

$$\begin{aligned} (a) \quad & \langle \xi - \omega'(z) + \omega'(w), u - w \rangle \geq 0 \forall u \in Z, \\ (b) \quad & \langle \eta - \omega'(z) + \omega'(z_+), u - z_+ \rangle \geq 0 \forall u \in Z, \end{aligned}$$

whence for every $u \in Z$ we have

$$\begin{aligned} \langle \eta, w - u \rangle &= \langle \eta, z_+ - u \rangle + \langle \eta, w - z_+ \rangle \\ &\leq \langle \omega'(z) - \omega'(z_+), z_+ - u \rangle + \langle \eta, w - z_+ \rangle \quad [\text{by (b)}] \\ &= [\omega(u) - \omega(z) - \langle \omega'(z), u - z \rangle] - [\omega(u) - \omega(z_+) - \langle \omega'(z_+), u - z_+ \rangle] \\ &\quad + [\omega(z) - \omega(z_+) + \langle \omega'(z), z_+ - z \rangle + \langle \eta, w - z_+ \rangle] \\ &= V_z(u) - V_{z_+}(u) + [\langle \eta, w - z_+ \rangle - V_z(z_+)] \quad [\text{we have proved (5.5.77.a)}] \\ \delta &:= \langle \eta, w - z_+ \rangle - V_z(z_+) = \langle \eta - \xi, w - z_+ \rangle - V_z(z_+) + \langle \xi, w - z_+ \rangle \\ &\leq \langle \eta - \xi, w - z_+ \rangle - V_z(z_+) + \langle \omega'(z) - \omega'(w), w - z_+ \rangle \quad [\text{by (a) with } u = z_+] \\ &= \langle \eta - \xi, w - z_+ \rangle + [\omega(z) - \omega(z_+) + \langle \omega'(z), z_+ - z \rangle + \langle \omega'(z) - \omega'(w), w - z_+ \rangle] \\ &= \langle \eta - \xi, w - z_+ \rangle - [[\omega(z) + \langle \omega'(z), w - z \rangle - \omega(w)] + [\omega(w) + \langle \omega'(w), z_+ - w \rangle - \omega(z_+)]] \\ &= \langle \eta - \xi, w - z_+ \rangle - V_w(w) - V_{z_+}(z_+) \quad [\text{we have proved (5.5.77.b)}] \\ &\leq \langle \eta - \xi, w - z_+ \rangle - \frac{1}{2} [\|w - z\|^2 + \|w - z_+\|^2] \quad [\text{by strong convexity of } \omega] \\ &\leq \|\eta - \xi\|_* \|w - z_+\| - \frac{1}{2} [\|w - z\|^2 + \|w - z_+\|^2] \\ &\leq \frac{1}{2} \|\eta - \xi\|_*^2 - \frac{1}{2} \|w - z\|^2 \quad [\text{since } \|\eta - \xi\|_* \|w - z_+\| - \frac{1}{2} \|w - z_+\|^2 \leq \frac{1}{2} \|\eta - \xi\|_*^2], \end{aligned}$$

and we have arrived at (5.5.77.c)²¹. Thus, (5.5.77) is proved.

Relation (5.5.78) follows from the first inequality in (5.5.77) in exactly the same fashion as (5.3.50) was derived from (5.3.49) when proving Theorem 5.3.4.

All remaining claims in (i) and (ii) are immediate consequences of (5.5.77) and (5.5.78). \square

Note that (5.5.79) corresponds to $O(1/\epsilon)$ upper complexity bound.

5.5.4.4 Refinement

Same as for various versions of MD, the complexity analysis of MP can be refined. The corresponding version of Theorem 5.5.4 is as follows.

Theorem 5.5.5 *Let (5.3.53) take place, and let positive reals $\lambda_1, \lambda_2, \dots$ satisfy (5.3.28). Let convex-concave saddle point problem (SP) be solved by the MP algorithm, and let*

$$z^t = \frac{\sum_{\tau=1}^t \lambda_\tau w_\tau}{\sum_{\tau=1}^t \lambda_\tau}.$$

Then $z^t \in Z$, and for $t = 1, 2, \dots$ one has

$$\epsilon_{\text{sad}}(z^t) \leq \frac{\frac{\lambda_t}{\gamma_t} \frac{\Omega_*^2}{2} + \sum_{\tau=1}^t \frac{\lambda_\tau}{\gamma_\tau} \delta_\tau}{\sum_{\tau=1}^t \lambda_\tau}, \quad (5.5.81)$$

²¹Note that the derivation we have just carried out is based solely on (5.5.80) and is completely independent of what are ξ and η ; this observation underlies numerous extensions and modifications of MP.

with δ_τ defined by (5.5.77.a).

In particular, let $\gamma_\tau \geq \frac{1}{L}$ be such that $\delta_\tau \leq 0$ for all τ (as is the case when $\gamma_\tau \equiv \frac{1}{L}$ and (5.5.75) takes place, see (5.5.77.c)). Then

$$\epsilon_{\text{sad}}(z^t) \leq \frac{\frac{\lambda_t}{\gamma_t} \Omega_*^2}{2 \sum_{\tau=1}^t \lambda_\tau}. \quad (5.5.82)$$

Proof. Invoking Theorem 5.5.4, we get (5.5.77). This relation implies for every τ :

$$\forall (u \in Z) : \lambda_\tau \langle F(w_\tau), w_\tau - u \rangle \leq \frac{\lambda_\tau}{\gamma_\tau} [V_{z_\tau}(u) - V_{z_{\tau+1}}(u)] + \frac{\lambda_\tau}{\gamma_\tau} \delta_\tau,$$

whence, same as in the proof of Theorem 5.3.2, for $t = 1, 2, \dots$ we get

$$\max_{u \in Z} \Lambda_t^{-1} \sum_{\tau=1}^t \lambda_\tau \langle F(w_\tau), w_\tau - u \rangle \leq \frac{\frac{\lambda_t}{\gamma_t} \Omega_*^2 + \sum_{\tau=1}^t \frac{\lambda_\tau}{\gamma_\tau} \delta_\tau}{\Lambda_t},$$

$$\Lambda_t = \sum_{\tau=1}^t \lambda_\tau.$$

Same as in the proof of Theorem 5.3.4, the latter inequality implies (5.5.81). The latter relation under the premise of Theorem immediately implies (5.5.82). \square

Incorporating aggressive stepsize policy. The efficiency estimate (5.5.79) in Theorem 5.5.4 is the better the larger are the stepsizes γ_τ ; however, this bound is obtained under the assumption that we maintain the relation $\delta_\tau \leq 0$, and the proof of Theorem suggests that when the stepsizes are greater than the “theoretically safe” value $\gamma_{\text{safe}} = \frac{1}{L}$, there are no guarantees for δ_τ to stay nonnegative. This does not mean, however, that all we can do is to use the worst-case-oriented stepsize policy $\gamma_\tau = \gamma_{\text{safe}}$. A reasonable “aggressive” stepsize policy is as follows: we start iteration τ with a “proposed value” $\gamma_\tau^+ \geq \gamma_{\text{safe}}$ of γ_τ and use $\gamma_\tau = \gamma_\tau^+$ to compute candidate w_τ , $z_{\tau+1}$ and the corresponding δ_τ . If $\delta_\tau \leq 0$, we consider our trial successful, treat the candidate w_τ and $z_{\tau+1}$ as actual ones and pass to iteration $\tau + 1$, increasing the value of γ^+ by an absolute constant factor: $\gamma_{\tau+1}^+ = \theta^+ \gamma_\tau^+$, where $\theta^+ > 1$ is a parameter of our strategy. On the other hand, if with $\gamma_\tau = \gamma_\tau^+$ we get $\delta_\tau > 0$, we consider the trial unsuccessful, reset γ_τ to a smaller value $\max[\theta_- \gamma_\tau, \gamma_{\text{safe}}]$ ($\theta_- \in (0, 1)$ is another parameter of our stepsize policy) and recompute the candidate w_τ and $z_{\tau+1}$ with γ_τ set to the new, reduced value. If we still get $\delta_\tau > 0$, we once again reduce the value of γ_τ and rerun the iteration, proceeding in this fashion until $\delta_\tau \leq 0$ is obtained. When it happens, we pass to iteration $\tau + 1$, now setting $\gamma_{\tau+1}^+$ to the value of γ_τ we have reached.

Computational experience shows that with reasonable choice of θ_\pm (our usual choice is $\theta_+ = 1.2$, $\theta_- = 0.8$) the aggressive policy outperforms the theoretically safe (i.e., with $\gamma_\tau \equiv \gamma_{\text{safe}}$) one: for the former policy, the “step per oracle call” index defined as the ratio of the quantity $\Gamma_t = \sum_{\tau=1}^T \gamma_\tau$ participating in the efficiency estimate (5.5.79) to the total number of computations of $F(\cdot)$ in course of T steps (including computations of F at the finally discarded trial intermediate points), where T is the step where we terminate the solution process, usually is essentially larger (for some problems – larger by orders of magnitude) than the similar ratio (i.e., $\gamma_{\text{safe}}/2$) for the safe stepsize policy.

5.5.4.5 Typical implementation

In a typical implementation of the MP algorithm, the components X and Y of the domain of the saddle problem of interest are convex compact subsets of direct products of finitely many *standard blocks*, specifically, Euclidean, ℓ_1/ℓ_2 and nuclear norm balls. For example, the usual box is the direct product of several segments (and thus – of one-dimensional Euclidean balls), a (standard) simplex is a subset of the ℓ_1 ball (that is, of a specific ℓ_1/ℓ_2 ball)²², a spectahedron is a subset of a specific nuclear norm ball, etc. In the sequel, we assume that X and Y are of the same dimensions as the direct products of standard blocks embedding these two sets. In the case in question, Z is a subset of the direct product $\widehat{Z} = Z_1 \times \dots \times Z_N \subset E = E_1 \times \dots \times E_N$ (E_j are the Euclidean spaces embedding Z_j) of standard blocks Z_j and the dimension of Z is equal to the one of \widehat{Z} . Recall that we have associated standard blocks and their embedding spaces with standard proximal setups, see section 5.3.3. The most natural way to get a proximal setup for the saddle point problem in question is to aggregate the proximal setups for separate blocks into a proximal setup $(\omega(\cdot), \|\cdot\|)$ for $\widehat{Z} \subset E$, and then to “reduce” this setup onto Z , meaning that we keep the norm $\|\cdot\|$ on the embedding space E (this space is common for $Z \subset \widehat{Z}$ and \widehat{Z}) intact and restrict the DGF ω from \widehat{Z} onto Z . The latter is legitimate: since Z and $\widehat{Z} \supset Z$ are of the same dimension, $\omega|_Z$ is a DGF for Z compatible with $\|\cdot\|$, and the $\omega|_Z$ -radius of Z clearly does not exceed the quantity

$$\widehat{\Omega} = \sqrt{2\Theta}, \quad \Theta = \max_{\widehat{Z}} \omega(\cdot) - \min_{\widehat{Z}} \omega(\cdot).$$

Now, the simplest way to aggregate the standard proximal setups for the standard blocks Z_j into such a setup for \widehat{Z} is as follows: denoting by $\|\cdot\|_{(j)}$ the norms, and by $\omega_j(z_j)$ – the DGFs as given by setups for blocks, we equip the embedding space of \widehat{Z} with the norm

$$\|[z^1; \dots; z^N]\| = \sqrt{\sum_{j=1}^N \alpha_j \|z^j\|_{(j)}^2},$$

and equip \widehat{Z} with the DGF

$$\omega(z^1, \dots, z^N) = \sum_{j=1}^N \beta_j \omega_j(z^j).$$

Here α_j and β_j are positive aggregation weights, which we choose in order to end up with a legitimate proximal setup for $Z \subset E$ with the best possible efficiency estimate. The standard recipe is as follows (for justification, see [21]):

Our initial data are the “partial proximal setups” $\|\cdot\|_{(j)}$, $\omega_j(\cdot)$ along with the upper bounds Θ_j on the variations

$$\max_{Z_j} \omega_j(\cdot) - \min_{Z_j} \omega_j(\cdot).$$

Now, the direct product structure of the embedding space of Z allows to represent the vector field F associated with the convex-concave function in question in the “block form”

$$F(z^1, \dots, z^N) = [F_1(z); \dots; F_N(z)].$$

²²We could treat the standard simplex as a standard block associated with the Entropy/Simplex proximal setup; to save words, we prefer to treat standard simplexes as parts of the unit ℓ_1 balls.

Assuming this field Lipschitz continuous, we can find (upper bounds on) the “partial Lipschitz constants” L_{ij} of this field, so that

$$\forall(i, z = [z^1; \dots; z^N] \in Z, w = [w^1; \dots; w^n] \in Z) : \|F_i(z) - F_i(w)\|_{(i,*)} \leq \sum_{j=1}^N \|z^j - w^j\|_{(j)},$$

where $\|\cdot\|_{(i,*)}$ are the norms conjugate to $\|\cdot\|_{(i)}$. W.l.o.g. we assume the bounds L_{ij} symmetric: $L_{ij} = L_{ji}$.

In terms of the parameters of the partial setups and the quantities L_{ij} , the aggregation is as follows: we set

$$\begin{aligned} M_{ij} &= L_{ij}\sqrt{\Theta_i\Theta_j}, \quad \sigma_k = \frac{\sum_j M_{kj}}{\sum_{i,j} M_{ij}}, \quad \alpha_j = \frac{\sigma_j}{\Theta_j}, \\ \|z^1; \dots; z^N\|^2 &= \sum_{j=1}^N \alpha_j \|z^j\|_{(j)}^2, \quad \omega(z^1, \dots, z^j) = \sum_{j=1}^N \alpha_j \omega_j(z^j). \end{aligned}$$

which results in

$$\Theta = \max_{\widehat{Z}} \omega(\cdot) - \min_{\widehat{Z}} \omega(\cdot) \leq 1,$$

(that is, the ω -radius of Z does not exceed $\sqrt{2}$). The condition (5.5.75) is now satisfied with

$$L = \mathcal{L} := \sum_{i,j} L_{ij} \sqrt{\Theta_i\Theta_j},$$

and the efficiency estimate (5.5.79) becomes

$$\epsilon_{\text{sad}}(z^t) \leq \mathcal{L} t^{-1}, \quad t = 1, 2, \dots$$

Illustration. Let us look at the problem

$$\text{Opt}_{r,p} = \min_{x \in X = \{x \in \mathbf{R}^n : \|x\|_p \leq 1\}} \|Bx - b\|_r,$$

where B is an $m \times n$ matrix and $p \in \{1, 2\}$, $r \in \{2, \infty\}$.

The problem admits an immediate saddle point reformulation:

$$\min_{x \in X = \{x \in \mathbf{R}^n : \|x\|_p \leq 1\}} \max_{y \in Y = \{y \in \mathbf{R}^m : \|y\|_{r_*} \leq 1\}} y^T(Bx - b)$$

where, as always,

$$r_* = \frac{r}{r-1}.$$

We are in the situation of $Z_1 = X$, $Z_2 = Y$ with

$$F(z^1 \equiv x, z^2 \equiv y) = [F_1(y) = B^T y; F_2(x) = b - Bx].$$

Besides this, the blocks Z_1 , Z_2 are either unit $\|\cdot\|_2$ -balls, or unit ℓ_1 balls. The associated quantities Θ_i , L_{ij} , $i, j = 1, 2$, within absolute constant factors are as follows:

$$\begin{array}{c|cc} & p=1 & p=2 \\ \hline \Theta_1 & \ln(n) & 1 \end{array} \quad \begin{array}{c|cc} & r=2 & r=\infty \\ \hline \Theta_2 & 1 & \ln(m) \end{array}$$

$$[L_{ij}]_{i,j=1}^2 = \begin{bmatrix} 0 & \|B\|_{p \rightarrow r} \\ \|B\|_{p \rightarrow r} & 0 \end{bmatrix}, \quad \|B\|_{p \rightarrow r} = \max_{\|x\|_p \leq 1} \|Bx\|_r.$$

The matrix norms $\|B\|_{p \rightarrow \max[r, 2]}$, again within an absolute constant factors, are given in the following table:

	$p = 1$	$p = 2$
$r = 1$	$\max_{j \leq n} \ \text{Col}_j[B]\ _2$	$\sigma_{\max}(B)$
$r = 2$	$\max_{j \leq n} \ \text{Col}_j[B]\ _2$	$\sigma_{\max}(B)$

where, as always, $\text{Col}_j[B]$ is j -th column of matrix B . For example, consider the situations of extreme interest for Compressed Sensing, namely, ℓ_1 -minimization with $\|\cdot\|_2$ -fit and ℓ_1 minimization with ℓ_∞ -fit (that is, the cases $(p = 1, r = 2)$ and $(p = 1, r = \infty)$, respectively). Here the achievable for MP efficiency estimates (the best known so far in the large scale case, although achievable not only with MP) are as follows:

- ℓ_1 -minimization with $\|\cdot\|_2$ -fit:

$$\|x^t\|_1 \leq 1 \& \|Bx^t - b\|_2 \leq \min_{\|x\|_1 \leq 1} \|Bx - x\|_2 + O(1) \frac{\sqrt{\ln(n)} \max_j \|\text{Col}_j[B]\|_2}{t};$$

- ℓ_1 -minimization with $\|\cdot\|_\infty$ -fit:

$$\|x^t\|_1 \leq 1 \& \|Bx^t - b\|_\infty \leq \min_{\|x\|_1 \leq 1} \|Bx - x\|_\infty + O(1) \frac{\sqrt{\ln(m) \ln(n)} \max_{i,j} |B_{ij}|}{t};$$

5.6 Appendix: Some proofs

5.6.1 Proofs of Facts 5.3.1, 5.3.2

5.6.1.1 A useful technical lemma

Lemma 5.6.1 *Let X be a closed convex domain in Euclidean space E , $(\|\cdot\|, \omega(\cdot))$ be a proximal setup for X , let $\Psi : X \rightarrow \mathbf{R} \cup \{+\infty\}$ be a lower semicontinuous convex function which is finite on the relative interior of X , and let $\phi(\cdot) : X \rightarrow \mathbf{R}$ be a convex continuously differentiable function. Assume that z_+ is a minimizer of the function $\omega(w) + \phi(w) + \Psi(w)$ on X . Then $z_+ \in X^\circ$, and there exists $\Psi' \in \partial\Psi(z_+)$ such that $\langle \omega'(z_+) + \nabla\phi(z_+) + \Psi', w - z_+ \rangle \geq 0$ for all $w \in X$. If $\Psi(\cdot)$ is differentiable at z_+ , one can take $\Psi' = \nabla\Psi(z_+)$.*

Proof. Let us set $\widehat{\omega}(w) = \omega(w) + \phi(w)$. Since ϕ is continuously differentiable on X , $\widehat{\omega}$ possesses the same smoothness properties as ω , specifically, $\widehat{\omega}$ is convex and continuous on X , $\{w \in X : \partial\widehat{\omega}(w) \neq \emptyset\} = X^\circ := \{w : \partial\omega(w) \neq \emptyset\}$ and $\widehat{\omega}'(w) := \omega'(w) + \nabla\phi(w) : X^\circ \rightarrow E$ is a continuous selection of subgradients of $\widehat{\omega}$.

W.l.o.g. we can assume that $\text{int } X \neq \emptyset$ and that $z_+ = 0$; note that z_+ is a minimizer of $\widehat{\omega}(w) + \Psi(w)$ over $w \in X$. Let $\bar{w} \in \text{int } X$, and let $w_t = t\bar{w}$, so that $w_t \in \text{int } X$ for $0 < t \leq 1$.

Let us prove first that $z_+ = 0 \in X^\circ$. To this end it suffices to verify that the vectors $\widehat{\omega}'(w_t)$ remain bounded as $t \rightarrow +0$; indeed, in this case the set of limiting points of $\widehat{\omega}'(w_t)$ as $t \rightarrow +0$ is nonempty; since $\widehat{\omega}(\cdot)$ is continuous on X , every one of these limiting points belongs to $\partial\widehat{\omega}(0)$, and thus $\partial\widehat{\omega}(0) \neq \emptyset$, meaning that $0 \in X^\circ$. To prove that $\widehat{\omega}'(w_t)$ is bounded as $t \rightarrow +0$, let $r > 0$ be such that with $B = \{h \in E : \|h\|_E \leq r\}$ we have $B^+ = \bar{w} + B \subset \text{int } X$. All we need to prove is that the linear forms $\langle \widehat{\omega}'(w_t), h \rangle$ of $h \in E$ are uniformly in $t \in (0, 1/2)$ bounded from above on B . Convexity of $\widehat{\omega}$ says that these forms are uniformly bounded from

above on B^+ , since with $\|h\|_E \leq r$, we have $\langle \widehat{\omega}'(w_t), \bar{w} + h \rangle = (1-t)^{-1} \langle \widehat{\omega}'(w_t), [\bar{w} + (1-t)h] - w_t \rangle \leq (1-t)^{-1} [\max_{w \in B^+} \widehat{\omega}(w) - \min_{0 \leq t \leq 1} \widehat{\omega}(w_t)] < \infty$. All we need to derive from uniform in $0 < t \leq 1/2$ above boundedness of the linear forms $f_t(h) = \langle \widehat{\omega}'(w_t), h \rangle$ on B^+ the desired uniform in t above boundedness of these forms on B is to verify that the quantities $\langle \widehat{\omega}'(w_t), \bar{w} \rangle$ are uniformly in $0 < t \leq 1/2$ below bounded. Assume, on the contrary, that for some sequence of $t_i > 0$ converging to 0 as $i \rightarrow \infty$ we have $p_i := \langle \widehat{\omega}'(w_{t_i}), \bar{w} \rangle \rightarrow -\infty$, and let $q = \langle \Psi'(\bar{w}), \bar{w} \rangle$ with $\Psi'(\bar{w}) \in \partial\Psi(\bar{w})$. Then by convexity of Ψ and due to $w_t = t\bar{w}$ we have $\Psi(0) \geq \Psi(w_t) - qt$, and by convexity of $\widehat{\omega}(\cdot)$ we have $\widehat{\omega}(0) \geq \widehat{\omega}(w_t) - t\langle \widehat{\omega}'(w_t), \bar{w} \rangle$, so that $0 \geq \widehat{\omega}(0) + \Psi(0) - [\widehat{\omega}(w_{t_i}) + \Psi(w_{t_i})] \geq -t_i(p_i + q)$ (recall that $z_+ = 0$ is the minimizer of $\widehat{\omega} + \Psi$ on X) for all i , which contradicts $p_i \rightarrow -\infty$, $i \rightarrow \infty$.

Now, since $0 = z_+ \in \text{Dom } \Psi$ and $\text{Dom } \Psi$ contains the interior of X , from lower semicontinuity and convexity of Ψ on the segment $[0, \bar{w}]$ it follows that Ψ is continuous on this segment; setting $\psi(t) = \Psi(w_t)$, $\phi(t) = \widehat{\omega}(w_t)$, observe that from the smoothness properties of $\widehat{\omega}(\cdot)$ it follows that ϕ is continuously differentiable on $[0, 1]$ with the derivative $\phi'(t) = \langle \widehat{\omega}'(w_t), \bar{w} \rangle$; since $\psi(t) + \phi(t)$ attains its minimum on $[0, 1]$ at $t = 0$, it immediately follows that the right derivative $\psi'(0)$ of ψ at $t = 0$ is $\geq -\phi'(0) = -\langle \widehat{\omega}'(0), \bar{w} \rangle$, whence $\Psi(\bar{w}) - \Psi(0) = \psi(1) - \psi(0) \geq \langle -\widehat{\omega}'(0), \bar{w} \rangle$. The relation $\Psi(\bar{w}) - \Psi(0) \geq -\langle \widehat{\omega}'(0), \bar{w} \rangle$ holds true for all $\bar{w} \in \text{int } X$ and therefore, by convexity of Ψ , for all $w \in X$, meaning that $\Psi' := -\widehat{\omega}'(0) \in \partial\Psi(0)$, and of course $\langle \Psi' + \widehat{\omega}'(0), w \rangle \geq 0$ for all w . Finally, when $\Psi(\cdot)$ is differentiable at $z_+ = 0$, we have $\psi'(0) = \langle \nabla\Psi(0), \bar{w} \rangle$; since we have already seen that $\psi''(0) + \phi'(0) \geq 0$, we get $\langle \nabla\Psi(0) + \widehat{\omega}'(0), \bar{w} \rangle \geq 0$. This inequality holds true for all $\bar{w} \in \text{int } X$ and therefore - for all $\bar{w} \in X$. Recalling that $\widehat{\omega}'(0) = \omega'(0) + \nabla\phi(0)$, the proof is complete. \square

5.6.1.2 Proof of Fact 5.3.1

If \bar{x} is a minimizer of $\omega(x) + \langle p, x \rangle$, then $-p$ clearly is a subgradient of $\omega(\cdot)$ at x , so that $\bar{x} \in X^\circ$. Vice versa, an $\bar{x} \in X^\circ$ is $\text{argmin}_{x \in X} \{\langle p, x \rangle + \omega(x)\}$ when $b = -\omega'(\bar{x})$. Further, if (5.3.5) takes place, then clearly \bar{x} is the minimizer of $\omega(x) + \langle p, x \rangle$ over X . Now let \bar{x} be a minimizer of $f(x) = \langle p, x \rangle + \omega(x)$ over X . Applying Lemma 5.6.1 with $\Psi(x) \equiv 0$, $\phi(x) = \langle p, x \rangle$, we see that $\bar{x} \in X^\circ$ and $\langle \omega'(\bar{x}) + p, x - \bar{x} \rangle \geq 0$ for all $x \in X$. \square

5.6.1.3 Proof of Fact 5.3.2

In one direction the statement is evident: if $x_* \in U \cap X^\circ$ satisfies (5.3.6), then clearly x_* minimizes $h_p(\cdot)$ on U . Further, the minimizer of h_p on U clearly exists and is unique due to the strong convexity of $h_p(\cdot)$. Thus, all we need to prove is that the minimizer x_* of h_p over U belongs to X° and satisfies (5.3.6). The demonstration is as follows: we can assume w.l.o.g. that X spans the entire \mathbf{R}^n and thus its relative interior is the same as its interior. Let $y \in U \cap \text{int } X$, and let $x_t = x_* + t(y - x_*)$, so that $x_t \in U \cap \text{int } X \subset X^\circ$ for $0 < t \leq 1$. Consequently, the function $\phi(t) = h_p(x_t)$ is convex, continuously differentiable on $(0, 1]$ with the derivative $\phi'(t) = \langle \omega'(x_t) + p, y - x_* \rangle$, and attains its minimum on $[0, 1]$ at $t = 0$, whence $\phi'(t) \geq 0$ for $0 < t \leq 1$. Now let $r > 0$ be such that the Euclidean ball B of radius r centered at y is contained in X . Since h_p is continuous on X , we have $h_p(z) - h_p(x_t) \leq V < \infty$ for all $z \in B$ and all

$t \in [0, 1]$, whence $\langle p + \omega'(x_t), z - x_t \rangle \leq V$ for all $t \in (0, 1]$ and all $z \in B$. On the other hand,

$$\begin{aligned} V &\geq \max_{z \in B} \langle p + \omega'(x_t), z - x_t \rangle = \langle p + \omega'(x_t), y - x_t \rangle + r\|p + \omega'(x_t)\|_2 \\ &= \phi'(t) + r\|p + \omega'(x_t)\|_2 \geq r\|p + \omega'(x_t)\|_2. \end{aligned}$$

We see that $\omega'(x_t)$ remains bounded as $t \rightarrow 0$; this, there exists a sequence $t_i \rightarrow +0$, $i \rightarrow \infty$, and $e \in \mathbf{R}^n$ such that $\omega'(x_{t_i}) \rightarrow e$ as $i \rightarrow \infty$. Since ω is continuous on X and $x_{t_i} \rightarrow x_*$ as $i \rightarrow \infty$, $e \in \partial\omega(x_*)$, that is, $x_* \in X^o$, whence in fact $\omega'(x_t) \rightarrow \omega'(x_*)$, $t \rightarrow +0$, and thus $\langle p + \omega'(x_*), y - x_* \rangle = \lim_{t \rightarrow +0} \phi'(t) \geq 0$. Thus, $x_* \in X^o$ and (5.3.6) holds true for all $u \in U \cap \text{int } X$. The latter set is dense in U , so that (5.3.6) holds true for all $u \in U$, as claimed. \square

5.6.2 Justifying Ball setup

Justification of Ball setup is trivial.

5.6.3 Justifying Entropy setup

When justifying Entropy setup, the only not completely trivial task is to check that $\omega(x = \sum_{i=1}^n x_i \ln(x_i))$ is strongly convex, modulus 1, w.r.t. $\|\cdot\|_1$, on the set $\Delta^o = \{x \in \Delta_n^+ : x > 0\}$. An evident sufficient condition for this is the relation

$$D^2(x)[h, h] := \frac{d^2}{dt^2}|_{t=0} \omega(x + th) \geq \kappa \|h\|^2 \quad \forall (x, h : x > 0, \sum_i x_i \leq 1) \quad (5.6.1)$$

(why?) To verify (5.6.1), observe that for $x \in \Delta^o$ we have

$$\|h\|_1^2 = \left[\sum_i |h_i| \right]^2 = \left[\sum_i \frac{|h_i|}{\sqrt{x_i}} \sqrt{x_i} \right]^2 \leq \left[\sum_i x_i \right] \left[\sum_i \frac{h_i^2}{x_i} \right] = h^T \omega''(x) h.$$

5.6.4 Justifying ℓ_1/ℓ_2 setup

All we need is given by the following two claims:

Theorem 5.6.1 *Let $E = \mathbf{R}^{k_1} \times \dots \times \mathbf{R}^{k_n}$, $\|\cdot\|$ be the ℓ_1/ℓ_2 norm on E , and let Z be the unit ball of the norm $\|\cdot\|$. Then the function*

$$\omega(x = [x^1; \dots; x^n]) = \frac{1}{p\gamma} \sum_{j=1}^n \|x^j\|_2^p, \quad p = \begin{cases} 2, & n \leq 2 \\ 1 + \frac{1}{\ln n}, & n \geq 3 \end{cases}, \quad \gamma = \begin{cases} 1, & n = 1 \\ 1/2, & n = 2 \\ \frac{1}{e \ln(n)}, & n > 2 \end{cases} : Z \rightarrow \mathbf{R} \quad (5.6.2)$$

is a continuously differentiable DGF for Z compatible with $\|\cdot\|$. As a result, whenever $X \subset Z$ is a nonempty closed convex subset of Z , the restriction $\omega|_X$ of ω on X is DGF for X compatible with $\|\cdot\|$, and the $\omega|_X$ -radius of X can be bounded as

$$\Omega \leq \sqrt{2e \ln(n+1)}. \quad (5.6.3)$$

Corollary 5.6.1 Let E and $\|\cdot\|$ be as in Theorem 5.6.1. Then the function

$$\widehat{\omega}(x = [x^1; \dots; x^n]) = \frac{n^{(p-1)(2-p)/p}}{2\gamma} \left[\sum_{j=1}^n \|x^j\|_2^p \right]^{\frac{2}{p}} : E \rightarrow \mathbf{R} \quad (5.6.4)$$

with p, γ given by (5.6.2) is a continuously differentiable DGF for E compatible with $\|\cdot\|$. As a result, whenever X is a nonempty closed subset of E , the restriction $\widehat{\omega}|_X$ of $\widehat{\omega}$ onto X is a DGF for X compatible with $\|\cdot\|$. When, in addition, $X \subset \{x \in E : \|x\| \leq R\}$ with $R < \infty$, the $\widehat{\omega}|_X$ -radius of X can be bounded as

$$\Omega \leq O(1) \sqrt{\ln(n+1)} R. \quad (5.6.5)$$

5.6.4.1 Proof of Theorem 5.6.1

We have for $x = [x^1; \dots; x^n] \in Z' = \{x \in Z : x^j[x] \neq 0 \forall j\}$, and $h = [h^1; \dots; h^n] \in E$:

$$\begin{aligned} \gamma D\omega(x)[h] &= \sum_{j=1}^n \|x^j\|_2^{p-2} \langle x^j, h^j \rangle \\ \gamma D^2\omega(x)[h, h] &= -(2-p) \sum_{j=1}^n \|x^j\|_2^{p-4} [\langle x^j, h^j \rangle]^2 + \sum_{j=1}^n \|x^j\|_2^{p-2} \|h^j\|_2^2 \\ &\geq \sum_{j=1}^n \|x^j\|_2^{p-2} \|h^j\|_2^2 - (2-p) \sum_{j=1}^n \|x^j\|_2^{p-4} \|x^j\|_2^2 \|h^j\|_2^2 \\ &\geq (p-1) \sum_{j=1}^n \|x^j\|_2^{p-2} \|h^j\|_2^2 \\ &\Rightarrow \left[\sum_j \|h^j\|_2 \right]^2 = \left[\sum_{j=1}^n [\|h^j\|_2 \|x^j\|_2^{\frac{p-2}{2}}] \|x^j\|_2^{\frac{2-p}{2}} \right]^2 \leq \left[\sum_{j=1}^n \|h^j\|_2^2 \|z^j\|_2^{p-2} \right] \left[\sum_{j=1}^n \|x^j\|_2^{2-p} \right] \\ &\Rightarrow \left[\sum_j \|h^j\|_2 \right]^2 \leq \left[\sum_{j=1}^n \|x^j\|_2^{2-p} \right] \frac{\gamma}{p-1} D^2\omega(x)[h, h] \end{aligned}$$

Setting $t_j = \|x^j\|_2 \geq 0$, we have $\sum_j t_j \leq 1$, whence due to $0 \leq 2-p \leq 1$ it holds $\sum_j t_j^{2-p} \leq n^{p-1}$. Thus,

$$\left[\sum_j \|h^j\|_2 \right]^2 \leq n^{p-1} \frac{\gamma}{p-1} D^2\omega(x)[h, h] \quad (5.6.6)$$

while

$$\max_{x \in Z} \omega(x) - \min_{x \in Z} \omega(x) \leq \frac{1}{\gamma p} \quad (5.6.7)$$

With p, γ as stated in Theorem 5.6.1, when $n \geq 3$ we get $\frac{\gamma}{p-1} n^{p-1} = 1$, and similarly for $n = 1, 2$. Consequently,

$$\forall (x \in Z', h = [h^1; \dots; h^n]) : \left[\sum_{j=1}^n \|h^j\|_2 \right]^2 \leq D^2\omega(x)[h, h]. \quad (5.6.8)$$

Since $\omega(\cdot)$ is continuously differentiable and the complement of Z' in Z is the union of finitely many proper linear subspaces of E , (5.6.8) implies that ω is strongly convex on Z , modulus 1, w.r.t. the ℓ_1/ℓ_2 norm $\|\cdot\|$. Besides this, we have

$$\frac{1}{\gamma p} \left\{ \begin{array}{ll} = \frac{1}{2}, & n = 1 \\ = 1, & n = 2 \\ \leq e \ln(n), & n \geq 3 \end{array} \right\} \leq O(1) \ln(n+1).$$

which combines with (5.6.7) to imply (5.6.3). \square

5.6.4.2 Proof of Corollary 5.6.1

Let

$$w(x = [x^1; \dots; x^n]) = \frac{1}{\gamma p} \sum_{j=1}^n \|x^j\|_2^p : E := \mathbf{R}^{k_1} \times \dots \times \mathbf{R}^{k_n} \rightarrow \mathbf{R} \quad (5.6.9)$$

with γ, p given by (5.6.2); note that $p \in (1, 2]$. As we know from Theorem 5.6.1, the restriction of $w(\cdot)$ onto the unit ball of ℓ_1/ℓ_2 norm $\|\cdot\|$ is continuously differentiable and strongly convex, modulus 1, w.r.t. $\|\cdot\|$. Besides this, $w(\cdot)$ clearly is nonnegative and positively homogeneous of the degree p : $w(\lambda x) \equiv |\lambda|^p w(x)$. Note that the outlined properties clearly imply that $w(\cdot)$ is convex and continuously differentiable on the entire E . We need the following simple

Lemma 5.6.2 *Let H be a Euclidean space, $\|\cdot\|_H$ be a norm (not necessary the Euclidean one), on H , and let $Y_H = \{y \in H : \|y\|_H \leq 1\}$. Let, further, $\zeta : H \rightarrow \mathbf{R}$ be a nonnegative continuously differentiable function, absolutely homogeneous of a degree $p \in (1, 2]$, and strongly convex, modulus 1, w.r.t. $\|\cdot\|_H$, on Y_H . Then the function*

$$\widehat{\zeta}(y) = \zeta^{2/p}(y) : H \rightarrow \mathbf{R}$$

is continuously differentiable homogeneous of degree 2 and strongly convex, modulus

$$\beta = \frac{2}{p} \left[\min_{y: \|y\|_H=1} \zeta^{2/p-1}(y) \right],$$

w.r.t. $\|\cdot\|_H$, on the entire H .

Proof. The facts that $\widehat{\zeta}(\cdot)$ is continuously differentiable convex and homogeneous of degree 2 function on H is evident, and all we need is to verify that the function is strongly convex, modulus β , w.r.t $\|\cdot\|_H$:

$$\langle \widehat{\zeta}'(y) - \widehat{\zeta}'(y'), y - y' \rangle \geq \beta \|y - y'\|_H^2 \quad \forall y, y'. \quad (5.6.10)$$

It is immediately seen that for a continuously differentiable function $\widehat{\zeta}$, the target property is local: to establish (5.6.10), it suffices to verify that every $\bar{y} \in H \setminus \{0\}$ admits a neighbourhood U such that the inequality in (5.6.10) holds true for all $y, y' \in U$. By evident homogeneity reasons, in the case of homogeneous of degree 2 function $\widehat{\zeta}$, to establish the latter property for every $\bar{y} \in H \setminus \{0\}$ is the same as to prove it in the case of $\|\bar{y}\|_H = 1 - \epsilon$, for any fixed $\epsilon \in (0, 1/3)$. Thus, let us fix an $\epsilon \in (0, 1)$ and $\bar{y} \in H$ with $\|\bar{y}\|_H = 1 - \epsilon$, and let $U = \{y' : \|y' - \bar{y}\|_H \leq \epsilon/2\}$, so that U is a compact set contained in the interior of Y_H . Now let $\phi(\cdot)$ be a nonnegative C^∞ function on H which vanishes outside of Y_H and has unit integral w.r.t. the Lebesgue measure on H . For $0 < r \leq 1$, let us set

$$\phi_r(y) = r^{-\dim H} \phi(y/r), \quad \zeta_r(\cdot) = (\zeta * \phi_r)(\cdot), \quad \widehat{\zeta}_r(\cdot) = \zeta_r^{2/p}(\cdot),$$

where $*$ stands for convolution. As $r \rightarrow +0$, the nonnegative C^∞ function $\zeta_r(\cdot)$ converges with the first order derivative, uniformly on U , to $\zeta(\cdot)$, and $\widehat{\zeta}_r(\cdot)$ converges with the first order derivative, uniformly on U , to $\widehat{\zeta}(\cdot)$. Besides this, when $r < \epsilon/2$, the function $\zeta_r(\cdot)$ is, along with $\zeta(\cdot)$, strongly convex, modulus 1, w.r.t. $\|\cdot\|_H$, on U (since $\zeta'_r(\cdot)$ is the convolution of $\zeta'(\cdot)$ and $\phi_r(\cdot)$). Since ζ_r is C^∞ , we conclude that when $r < \epsilon/2$, we have

$$D^2 \zeta_r(y)[h, h] \geq \|h\|_H^2 \quad \forall (y \in U, h \in H),$$

whence also

$$\begin{aligned} D^2\widehat{\zeta}_r(y)[h, h] &= \frac{2}{p} \left(\frac{2}{p} - 1 \right) \zeta_r^{2/p-2}(y) [D\zeta_r(y)[h]]^2 + \frac{2}{p} \zeta_r^{2/p-1}(y) D^2\zeta_r[h, h] \\ &\geq \frac{2}{p} \zeta_r^{2/p-1}(y) \|h\|_H^2 \quad \forall (h \in H, y \in U). \end{aligned}$$

It follows that when $r < \epsilon/2$, we have

$$\begin{aligned} \langle \widehat{\zeta}'_r(y) - \widehat{\zeta}'_r(y'), y - y' \rangle &\geq \beta_r[U] \|y - y'\|_H^2 \quad \forall y, y' \in U, \\ \beta_r[U] &= \frac{2}{p} \min_{u \in U} \zeta_r^{2/p-1}(u). \end{aligned}$$

As $r \rightarrow +0$, $\widehat{\zeta}'_r(\cdot)$ uniformly on U converges to $\widehat{\zeta}'(\cdot)$, and $\beta_r[U]$ converges to $\beta[U] = \frac{2}{p} \min_{u \in U} \zeta^{2/p-1}(u) \geq (1 - 3\epsilon/2)^{2-p}\beta$ (recall the origin of β and U and note that ζ is absolutely homogeneous of degree p), and we arrive at the relation

$$\langle \widehat{\zeta}'(y) - \widehat{\zeta}'(y'), y - y' \rangle \geq (1 - 3\epsilon/2)^{2-p}\beta \|y - y'\|_H^2 \quad \forall y, y' \in U.$$

As we have already explained, the just established validity of the latter relation for every $U = \{y : \|y - \bar{y}\|_H \leq \epsilon/2\}$ and every \bar{y} , $\|\bar{y}\|_H = 1 - \epsilon$ implies that $\widehat{\zeta}$ is strongly convex, modulus $(1 - 3\epsilon/2)^{2-p}\beta$, on the entire H . Since $\epsilon > 0$ is arbitrary, Lemma 5.6.2 is proved. \square

B. Applying Lemma 5.6.2 to E in the role of H , the ℓ_1/ℓ_2 norm $\|\cdot\|$ in the role of $\|\cdot\|_H$ and the function $w(\cdot)$ given by (5.6.9) in the role of ζ , we conclude that the function $\widehat{w}(\cdot) \equiv w^{2/p}(\cdot)$ is strongly convex, modulus $\beta = \frac{2}{p} [\min_y \{w(y) : y \in E, \|y\| = 1\}]^{2/p-1}$, w.r.t. $\|\cdot\|$, on the entire space E , whence the function $\beta^{-1}w(z)$ is continuously differentiable and strongly convex, modulus 1, w.r.t. $\|\cdot\|$, on the entire E . This observation, in view of the evident relation $\beta = \frac{2}{p} (n^{1-p} \gamma^{-1} p^{-1})^{2/p-1}$ and the origin of γ, p , immediately implies all claims in Corollary 5.6.1. \square

5.6.5 Justifying Nuclear norm setup

All we need is given by the following two claims:

Theorem 5.6.2 *Let $\mu = (\mu_1, \dots, \mu_n)$ and $\nu = (n\nu_1, \dots, n\nu_n)$ be collections of positive reals such that $\mu_j \leq \nu_j$ for all j , and let*

$$m = \sum_{j=1}^n \mu_j.$$

With $E = \mathbf{M}^{\mu\nu}$ and $\|\cdot\| = \|\cdot\|_{\text{nuc}}$, let $Z = \{x \in E : \|x\|_{\text{nuc}} \leq 1\}$. Then the function

$$\omega(x) = \frac{4\sqrt{e} \ln(2m)}{2^q(1+q)} \sum_{i=1}^m \sigma_i^{1+q}(x) : Z \rightarrow \mathbf{R}, \quad q = \frac{1}{2 \ln(2m)}, \quad (5.6.11)$$

is continuously differentiable DGF for Z compatible with $\|\cdot\|_{\text{nuc}}$, so that for every nonempty closed convex subset X of Z , the function $\omega|_X$ is a DGF for X compatible with $\|\cdot\|_{\text{nuc}}$. Besides this, the $\omega|_X$ radius of X can be bounded as

$$\Omega \leq 2\sqrt{2\sqrt{e} \ln(2m)} \leq 4\sqrt{\ln(2m)}. \quad (5.6.12)$$

Corollary 5.6.2 Let $\mu, \nu, m, E, \|\cdot\|$ be as in Theorem 5.6.2. Then the function

$$\hat{\omega}(x) = 2e \ln(2m) \left[\sum_{j=1}^m \sigma_j^{1+q}(x) \right]^{\frac{2}{1+q}} : E \rightarrow \mathbf{R}, \quad q = \frac{1}{2 \ln(2m)} \quad (5.6.13)$$

is continuously differentiable DGF for $Z = E$ compatible with $\|\cdot\|_{\text{nuc}}$, so that for every nonempty closed convex set $X \subset E$, the function $\hat{\omega}|_X$ is a DGF for X compatible with $\|\cdot\|_{\text{nuc}}$. Besides this, if X is bounded, the $\hat{\omega}|_X$ radius of X can be bounded as

$$\Omega \leq O(1) \sqrt{\ln(2\mu)} R, \quad (5.6.14)$$

where $R < \infty$ is such that $X \subset \{x \in E : \|x\|_{\text{nuc}} \leq R\}$.

5.6.5.1 Proof of Theorem 5.6.2

A. As usual, let \mathbf{S}^n be the space of $n \times n$ symmetric matrices equipped with the Frobenius inner product; for $y \in \mathbf{S}^n$, let $\lambda(y)$ be the vector of eigenvalues of y (taken with their multiplicities in the non-ascending order), and let $|y|_1 = \|\lambda(y)\|_1$ be the trace norm.

Lemma 5.6.3 Let $N \geq M \geq 2$, and let F be a linear subspace in \mathbf{S}^N such that every matrix $y \in F$ has at most M nonzero eigenvalues. Let $q = \frac{1}{2 \ln(M)}$, so that $0 < q < 1$, and let

$$\hat{\omega}(y) = \frac{4\sqrt{e} \ln(M)}{1+q} \sum_{j=1}^N |\lambda_j(y)|^{1+q} : \mathbf{S}^N \rightarrow \mathbf{R}.$$

The function $\hat{\omega}(\cdot)$ is continuously differentiable, convex, and its restriction on the set $Y_F = \{y \in F : |y|_1 \leq 1\}$ is strongly convex, modulus 1, w.r.t. $|\cdot|_1$.

Proof:

1⁰. Let $0 < q < 1$. Consider the following function of $y \in \mathbf{S}^N$:

$$\chi(y) = \frac{1}{1+q} \sum_{j=1}^N |\lambda_j(y)|^{1+q} = \text{Tr}(f(y)), \quad f(s) = \frac{1}{1+q} |s|^{1+q}. \quad (5.6.15)$$

2⁰. Function $f(s)$ is continuously differentiable on the axis and twice continuously differentiable outside of the origin; consequently, we can find a sequence of polynomials $f_k(s)$ converging, as $k \rightarrow \infty$, to f along with their first derivatives uniformly on every compact subset of \mathbf{R} and, besides this, converging to f uniformly along with the first and the second derivative on every compact subset of $\mathbf{R} \setminus \{0\}$. Now let $y, h \in \mathbf{S}^N$, let $y = u \text{Diag}\{\lambda\} u^T$ be the eigenvalue decomposition of y , and let $h = u \hat{h} u^T$. For a polynomial $p(s) = \sum_{k=0}^K p_k s^k$, setting $P(w) = \text{Tr}(\sum_{k=0}^K p_k w^k) : \mathbf{S}^N \rightarrow \mathbf{R}$, and denoting by γ a closed contour in \mathbf{C} encircling the spectrum of y , we have

- (a) $P(y) = \text{Tr}(p(y)) = \sum_{j=1}^N p(\lambda_j(y))$
- (b) $DP(y)[h] = \sum_{k=1}^K k p_k \text{Tr}(y^{k-1} h) = \text{Tr}(p'(y) h) = \sum_{j=1}^N p'(\lambda_j(y)) \hat{h}_{jj}$
- (c) $D^2 P(y)[h, h] = \frac{d}{dt} \Big|_{t=0} DP(y + th)[h] = \frac{d}{dt} \Big|_{t=0} \text{Tr}(p'(y + th) h)$
 $= \frac{d}{dt} \Big|_{t=0} \frac{1}{2\pi i} \oint_{\gamma} \text{Tr}(h(zI - (y + th))^{-1}) p'(z) dz = \frac{1}{2\pi i} \oint_{\gamma} \text{Tr}(h(zI - y)^{-1} h(zI - y)^{-1}) p'(z) dz$
 $= \frac{1}{2\pi i} \oint_{\gamma} \sum_{i,j=1}^N \hat{h}_{ij}^2 \frac{p'(z)}{(z - \lambda_i(y))(z - \lambda_j(y))} dz = \sum_{i,j=1}^N \hat{h}_{ij}^2 \Gamma_{ij},$
 $\Gamma_{ij} = \begin{cases} \frac{p'(\lambda_i(y)) - p'(\lambda_j(y))}{\lambda_i(y) - \lambda_j(y)}, & \lambda_i(y) \neq \lambda_j(y) \\ p''(\lambda_i(y)), & \lambda_i(y) = \lambda_j(y) \end{cases}$

We conclude from (a, b) that as $k \rightarrow \infty$, the real-valued polynomials $F_k(\cdot) = \text{Tr}(f_k(\cdot))$ on \mathbf{S}^N converge, along with their first order derivatives, uniformly on every bounded subset of \mathbf{S}^N , and the limit of the sequence, by (a), is exactly $\chi(\cdot)$. Thus, $\chi(\cdot)$ is continuously differentiable, and (b) says that

$$D\chi(y)[h] = \sum_{j=1}^N f'(\lambda_j(y))\hat{h}_{jj}. \quad (5.6.16)$$

Besides this, (a-c) say that if U is a closed convex set in \mathbf{S}^N which does not contain singular matrices, then $F_k(\cdot)$, as $k \rightarrow \infty$, converge along with the first and the second derivative uniformly on every compact subset of U , so that $\chi(\cdot)$ is twice continuously differentiable on U , and at every point $y \in U$ we have

$$D^2\chi(y)[h, h] = \sum_{i,j=1}^N \hat{h}_{ij}^2 \Gamma_{ij}, \quad \Gamma_{ij} = \begin{cases} \frac{f'(\lambda_i(y)) - f'(\lambda_j(y))}{\lambda_i(y) - \lambda_j(y)}, & \lambda_i(y) \neq \lambda_j(y) \\ f''(\lambda_i(y)), & \lambda_i(y) = \lambda_j(y) \end{cases} \quad (5.6.17)$$

and in particular $\chi(\cdot)$ is convex on U .

3⁰. We intend to prove that (i) $\chi(\cdot)$ is convex, and (ii) its restriction on the set Y_F is strongly convex, with certain modulus $\alpha > 0$, w.r.t. the trace norm $|\cdot|_1$. Since χ is continuously differentiable, all we need to prove (i) is to verify that

$$\langle \chi'(y') - \chi'(y''), y' - y'' \rangle \geq 0 \quad (*)$$

for a dense in $\mathbf{S}^N \times \mathbf{S}^N$ set of pairs (y', y'') , e.g., those with nonsingular $y' - y''$. For a pair of the latter type, the polynomial $q(t) = \text{Det}(y' + t(y'' - y'))$ of $t \in \mathbf{R}$ is not identically zero and thus has finitely many roots on $[0, 1]$. In other words, we can find finitely many points $t_0 = 0 < t_1 < \dots < t_n = 1$ such that all “matrix intervals” $\Delta_i = (y_i, y_{i+1})$, $y_k = y' + t_k(y'' - y')$, $1 \leq i \leq n - 1$, are comprised of nonsingular matrices. Therefore χ is convex on every closed segment contained in one of Δ_i 's, and since χ is continuously differentiable, (*) follows.

4⁰. Now let us prove that with properly defined $\alpha > 0$ one has

$$\langle \chi'(y') - \chi'(y''), y' - y'' \rangle \geq \alpha |y' - y''|_1^2 \quad \forall y', y'' \in Y_F \quad (5.6.18)$$

Let $\epsilon > 0$, and let Y^ϵ be a convex open in $Y = \{y : |y|_1 \leq 1\}$ neighbourhood of Y_F such that for all $y \in Y^\epsilon$ at most M eigenvalues of y are of magnitude $> \epsilon$. We intend to prove that for some $\alpha_\epsilon > 0$ one has

$$\langle \chi'(y') - \chi'(y''), y' - y'' \rangle \geq \alpha_\epsilon |y' - y''|_1^2 \quad \forall y', y'' \in Y^\epsilon. \quad (5.6.19)$$

Same as above, it suffices to verify this relation for a dense in $Y^\epsilon \times Y^\epsilon$ set of pairs $y', y'' \in Y^\epsilon$, e.g., for those pairs $y', y'' \in Y^\epsilon$ for which $y' - y''$ is nonsingular. Defining matrix intervals Δ_i as above and taking into account continuous differentiability of χ , it suffices to verify that if $y \in \Delta_i$ and $h = y' - y''$, then $D^2\chi(y)[h, h] \geq \alpha_\epsilon |h|_1^2$. To this end observe that by (5.6.17) all we have to prove is that

$$D^2\chi(y)[h, h] = \sum_{i,j=1}^N \hat{h}_{ij}^2 \Gamma_{ij} \geq \alpha_\epsilon |h|_1^2. \quad (\#)$$

5⁰. Setting $\lambda_j = \lambda_j(y)$, observe that $\lambda_i \neq 0$ for all i due to the origin of y . We claim that if $|\lambda_i| \geq |\lambda_j|$, then $\Gamma_{ij} \geq q|\lambda_i|^{q-1}$. Indeed, the latter relation definitely holds true when $\lambda_i = \lambda_j$. Now, if λ_i and λ_j are of the same sign, then $\Gamma_{ij} = \frac{|\lambda_i|^q - |\lambda_j|^q}{|\lambda_i| - |\lambda_j|} \geq q|\lambda_i|^{q-1}$, since the derivative of

the concave (recall that $0 < q \leq 1$) function t^q of $t > 0$ is positive and nonincreasing. If λ_i and λ_j are of different signs, then $\Gamma_{ij} = \frac{|\lambda_i|^q + |\lambda_j|^q}{|\lambda_i| + |\lambda_j|} \geq |\lambda_i|^{q-1}$ due to $|\lambda_j|^q \geq |\lambda_j||\lambda_i|^{q-1}$, and therefore $\Gamma_{ij} \geq q|\lambda_i|^{q-1}$. Thus, our claim is justified.

W.l.o.g. we can assume that the positive reals $\mu_i = |\lambda_i|$, $i = 1, \dots, N$, form a nondecreasing sequence, so that, by above, $\Gamma_{ij} \geq q\mu_j^{q-1}$ when $i \leq j$. Besides this, at most M of μ_j are $\geq \epsilon$, since $y', y'' \in Y^\epsilon$ and therefore $y \in Y^\epsilon$ by convexity of Y^ϵ . By the above,

$$D^2\chi(y)[h, h] \geq 2q \sum_{i < j \leq N} \hat{h}_{ij}^2 \mu_j^{q-1} + q \sum_{j=1}^N \hat{h}_{jj}^2 \mu_j^{q-1},$$

or, equivalently by symmetry of \hat{h} , if

$$h^j = \begin{bmatrix} & & \hat{h}_{1j} \\ & & \hat{h}_{2j} \\ & \vdots & \\ \hat{h}_{j1} & \hat{h}_{j2} & \dots & \hat{h}_{jj} \end{bmatrix}$$

and $H_j = \|h^j\|_{\text{Fro}}$ is the Frobenius norm of h^j , then

$$D^2\chi(y)[h, h] \geq q \sum_{j=1}^N H_j^2 \mu_j^{q-1}. \quad (5.6.20)$$

Observe that h^j are of rank ≤ 2 , whence $|h^j|_1 \leq \sqrt{2}\|h^j\|_{\text{Fro}} = \sqrt{2}H_j$, so that

$$\begin{aligned} |h|_1^2 = |\hat{h}|_1^2 &\leq \left(\sum_{j=1}^N |h^j|_1 \right)^2 \leq 2 \left(\sum_{j=1}^N H_j \right)^2 = 2 \left(\sum_j [H_j \mu_j^{(q-1)/2}] \mu_j^{(1-q)/2} \right)^2 \\ &\leq 2 \left(\sum_{j=1}^N H_j^2 \mu_j^{q-1} \right) \left(\sum_{j=1}^N \mu_j^{1-q} \right) \\ &\leq (2/q) D^2\chi(y)[h, h] \left(\sum_{j=1}^N \mu_j^{1-q} \right) \quad [\text{by (5.6.20)}] \\ &\leq (2/q) D^2\chi(y)[h, h] \left((N-M)\epsilon^{1-q} + [M^{-1} \sum_{j=N-M+1}^N \mu_j]^{1-q} M \right) \\ &\quad [\text{due to } 0 < q < 1 \text{ and } \mu_j \leq \epsilon, j \leq N-M] \\ &\leq (2/q) D^2\chi(y)[h, h] ((N-M)\epsilon^{1-q} + M^q) \quad [\text{since } \sum_j \mu_j \leq 1 \text{ due to } y \in Y^\epsilon \subset \{y : |y|_1 \leq 1\}], \end{aligned}$$

and we see that (#) holds true with $\alpha_\epsilon = \frac{q}{2[(N-M)\epsilon + M^q]}$. As a result, with the just defined α_ϵ relation (5.6.19) holds true, whence (5.6.18) is satisfied with $\alpha = \lim_{\epsilon \rightarrow +0} \alpha_\epsilon = \frac{q}{2}M^{-q}$, that is, χ is continuously differentiable convex function which is strongly convex, modulus α , w.r.t. $|\cdot|_1$, on Y_F . Recalling the definition of q , we see that $\widehat{\omega}(\cdot) = \alpha^{-1}\chi(\cdot)$, so that $\widehat{\omega}$ satisfies the conclusion of Lemma 5.6.3. \square

B. We are ready to prove Theorem 5.6.2. Under the premise and in the notation of Theorem, let

$$N = \sum_j \mu_j + \sum_j \nu_j, \quad M = 2 \sum_j \mu_j = 2m, \quad \mathcal{A}x = \frac{1}{2} \begin{bmatrix} & x \\ x^T & \end{bmatrix} \in \mathbf{S}^N \quad [x \in E = \mathbf{M}^{\mu\nu}]. \quad (5.6.21)$$

Observe that the image space F of \mathcal{A} is a linear subspace of \mathbf{S}^N , and that the eigenvalues of $y = \mathcal{A}x$ are the $2m$ reals $\pm\sigma_i(x)/2$, $1 \leq i \leq m$, and $N - M$ zeros, so that $\|x\|_{\text{nuc}} \equiv |\mathcal{A}x|_1$ and M, F satisfy the premise of Lemma 5.6.3. Setting

$$\omega(x) = \widehat{\omega}(\mathcal{A}x) = \frac{4\sqrt{e}\ln(2m)}{2^q(1+q)} \sum_{i=1}^m \sigma_i^{1+q}(x), \quad q = \frac{1}{2\ln(2m)},$$

and invoking Lemma 5.6.3, we see that ω is a convex continuously differentiable function on E which, due to the identity $\|x\|_{\text{nuc}} \equiv |\mathcal{A}x|_1$, is strongly convex, modulus 1, w.r.t. $\|\cdot\|_{\text{nuc}}$, on the $\|\cdot\|_{\text{nuc}}$ -unit ball Z of E . This function clearly satisfies (5.6.12). \square

5.6.5.2 Proof of Corollary 5.6.2

Corollary 5.6.2 is obtained from Theorem 5.6.2 in exactly the same fashion as Corollary 5.6.1 was derived from Theorem 5.6.1, see section 5.6.4.

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Appendix A

Prerequisites from Linear Algebra and Analysis

Regarded as mathematical entities, the objective and the constraints in a Mathematical Programming problem are functions of several real variables; therefore before entering the Optimization Theory and Methods, we need to recall several basic notions and facts about the spaces \mathbf{R}^n where these functions live, same as about the functions themselves. The reader is supposed to know most of the facts to follow, so he/she should not be surprised by a “cooking book” style which we intend to use in this Lecture.

A.1 Space \mathbf{R}^n : algebraic structure

Basically all events and constructions to be considered will take place in the *space \mathbf{R}^n of n -dimensional real vectors*. This space can be described as follows.

A.1.1 A point in \mathbf{R}^n

A *point* in \mathbf{R}^n (called also an *n -dimensional vector*) is an ordered collection $x = (x_1, \dots, x_n)$ of n reals, called the *coordinates*, or *components*, or *entries* of vector x ; the space \mathbf{R}^n itself is the set of all collections of this type.

A.1.2 Linear operations

\mathbf{R}^n is equipped with two *basic operations*:

- *Addition of vectors.* This operation takes on input two vectors $x = (x_1, \dots, x_n)$ and $y = (y_1, \dots, y_n)$ and produces from them a new vector

$$x + y = (x_1 + y_1, \dots, x_n + y_n)$$

with entries which are sums of the corresponding entries in x and in y .

- *Multiplication of vectors by reals.* This operation takes on input a real λ and an n -dimensional vector $x = (x_1, \dots, x_n)$ and produces from them a new vector

$$\lambda x = (\lambda x_1, \dots, \lambda x_n)$$

with entries which are λ times the entries of x .

The as far as addition and multiplication by reals are concerned, the arithmetic of \mathbf{R}^n inherits most of the common rules of Real Arithmetic, like $x + y = y + x$, $(x + y) + z = x + (y + z)$, $(\lambda + \mu)(x + y) = \lambda x + \mu x + \lambda y + \mu y$, $\lambda(\mu x) = (\lambda\mu)x$, etc.

A.1.3 Linear subspaces

Linear subspaces in \mathbf{R}^n are, by definition, nonempty subsets of \mathbf{R}^n which are closed with respect to addition of vectors and multiplication of vectors by reals:

$$L \subset \mathbf{R}^n \text{ is a linear subspace} \Leftrightarrow \begin{cases} L \neq \emptyset; \\ x, y \in L \Rightarrow x + y \in L; \\ x \in L, \lambda \in \mathbf{R} \Rightarrow \lambda x \in L. \end{cases}$$

A.1.3.A. Examples of linear subspaces:

1. The entire \mathbf{R}^n ;
2. The *trivial* subspace containing the single zero vector $0 = (0, \dots, 0)$ ¹⁾; (this vector/point is called also the origin)
3. The set $\{x \in \mathbf{R}^n : x_1 = 0\}$ of all vectors x with the first coordinate equal to zero.

The latter example admits a natural extension:

4. The set of all solutions to a *homogeneous* (i.e., with zero right hand side) system of linear equations

$$\left\{ x \in \mathbf{R}^n : \begin{array}{rcl} a_{11}x_1 + \dots + a_{1n}x_n & = & 0 \\ a_{21}x_1 + \dots + a_{2n}x_n & = & 0 \\ \dots & & \\ a_{m1}x_1 + \dots + a_{mn}x_n & = & 0 \end{array} \right\} \quad (\text{A.1.1})$$

always is a linear subspace in \mathbf{R}^n . This example is “generic”, that is, every linear subspace in \mathbf{R}^n is the solution set of a (finite) system of homogeneous linear equations, see Proposition A.3.6 below.

5. *Linear span of a set of vectors*. Given a nonempty set X of vectors, one can form a linear subspace $\text{Lin}(X)$, called the *linear span* of X ; this subspace consists of all vectors x which can be represented as *linear combinations* $\sum_{i=1}^N \lambda_i x_i$ of vectors from X (in $\sum_{i=1}^N \lambda_i x_i$, N is an arbitrary positive integer, λ_i are reals and x_i belong to X). Note that

Lin(X) is the smallest linear subspace which contains X : if L is a linear subspace such that $L \supset X$, then $L \supset \text{Lin}(X)$ (why?).

The “linear span” example also is generic:

Every linear subspace in \mathbf{R}^n is the linear span of an appropriately chosen finite set of vectors from \mathbf{R}^n .

(see Theorem A.1.2.(i) below).

¹⁾Pay attention to the notation: we use the same symbol 0 to denote the real zero and the n -dimensional vector with all coordinates equal to zero; these two zeros are not the same, and one should understand from the context (it always is very easy) which zero is meant.

A.1.3.B. Sums and intersections of linear subspaces. Let $\{L_\alpha\}_{\alpha \in I}$ be a family (finite or infinite) of linear subspaces of \mathbf{R}^n . From this family, one can build two sets:

1. The sum $\sum_{\alpha} L_\alpha$ of the subspaces L_α which consists of all vectors which can be represented as finite sums of vectors taken each from its own subspace of the family;
2. The intersection $\bigcap_{\alpha} L_\alpha$ of the subspaces from the family.

Theorem A.1.1 Let $\{L_\alpha\}_{\alpha \in I}$ be a family of linear subspaces of \mathbf{R}^n . Then

- (i) The sum $\sum_{\alpha} L_\alpha$ of the subspaces from the family is itself a linear subspace of \mathbf{R}^n ; it is the smallest of those subspaces of \mathbf{R}^n which contain every subspace from the family;
- (ii) The intersection $\bigcap_{\alpha} L_\alpha$ of the subspaces from the family is itself a linear subspace of \mathbf{R}^n ; it is the largest of those subspaces of \mathbf{R}^n which are contained in every subspace from the family.

A.1.4 Linear independence, bases, dimensions

A collection $X = \{x^1, \dots, x^N\}$ of vectors from \mathbf{R}^n is called *linearly independent*, if no nontrivial (i.e., with at least one nonzero coefficient) linear combination of vectors from X is zero.

Example of linearly independent set: the collection of n standard basic orths $e_1 = (1, 0, \dots, 0)$, $e_2 = (0, 1, 0, \dots, 0)$, ..., $e_n = (0, \dots, 0, 1)$.

Examples of linearly dependent sets: (1) $X = \{0\}$; (2) $X = \{e_1, e_1\}$; (3) $X = \{e_1, e_2, e_1 + e_2\}$.

A collection of vectors f^1, \dots, f^m is called a *basis* in \mathbf{R}^n , if

1. The collection is linearly independent;
2. Every vector from \mathbf{R}^n is a linear combination of vectors from the collection (i.e., $\text{Lin}\{f^1, \dots, f^m\} = \mathbf{R}^n$).

Example of a basis: The collection of standard basic orths e_1, \dots, e_n is a basis in \mathbf{R}^n .

Examples of non-bases: (1) The collection $\{e_2, \dots, e_n\}$. This collection is linearly independent, but not every vector is a linear combination of the vectors from the collection; (2) The collection $\{e_1, e_1, e_2, \dots, e_n\}$. Every vector is a linear combination of vectors from the collection, but the collection is not linearly independent.

Besides the bases of the entire \mathbf{R}^n , one can speak about the bases of linear subspaces:

A collection $\{f^1, \dots, f^m\}$ of vectors is called a *basis of a linear subspace* L , if

1. The collection is linearly independent,
2. $L = \text{Lin}\{f^1, \dots, f^m\}$, i.e., all vectors f^i belong to L , and every vector from L is a linear combination of the vectors f^1, \dots, f^m .

In order to avoid trivial remarks, it makes sense to agree once for ever that

An empty set of vectors is linearly independent, and an empty linear combination of vectors $\sum_{i \in \emptyset} \lambda_i x_i$ equals to zero.

With this convention, the trivial linear subspace $L = \{0\}$ also has a basis, specifically, an empty set of vectors.

Theorem A.1.2 (i) *Let L be a linear subspace of \mathbf{R}^n . Then L admits a (finite) basis, and all bases of L are comprised of the same number of vectors; this number is called the dimension of L and is denoted by $\dim(L)$.*

We have seen that \mathbf{R}^n admits a basis comprised of n elements (the standard basic orths). From (i) it follows that every basis of \mathbf{R}^n contains exactly n vectors, and the dimension of \mathbf{R}^n is n .

(ii) *The larger is a linear subspace of \mathbf{R}^n , the larger is its dimension: if $L \subset L'$ are linear subspaces of \mathbf{R}^n , then $\dim(L) \leq \dim(L')$, and the equality takes place if and only if $L = L'$.*

We have seen that the dimension of \mathbf{R}^n is n ; according to the above convention, the trivial linear subspace $\{0\}$ of \mathbf{R}^n admits an empty basis, so that its dimension is 0. Since $\{0\} \subset L \subset \mathbf{R}^n$ for every linear subspace L of \mathbf{R}^n , it follows from (ii) that the dimension of a linear subspace in \mathbf{R}^n is an integer between 0 and n .

(iii) *Let L be a linear subspace in \mathbf{R}^n . Then*

(iii.1) *Every linearly independent subset of vectors from L can be extended to a basis of L ;*

(iii.2) *From every spanning subset X for L – i.e., a set X such that $\text{Lin}(X) = L$ – one can extract a basis of L .*

It follows from (iii) that

- every linearly independent subset of L contains at most $\dim(L)$ vectors, and if it contains exactly $\dim(L)$ vectors, it is a basis of L ;
- every spanning set for L contains at least $\dim(L)$ vectors, and if it contains exactly $\dim(L)$ vectors, it is a basis of L .

(iv) *Let L be a linear subspace in \mathbf{R}^n , and f^1, \dots, f^m be a basis in L . Then every vector $x \in L$ admits exactly one representation*

$$x = \sum_{i=1}^m \lambda_i(x) f^i$$

as a linear combination of vectors from the basis, and the mapping

$$x \mapsto (\lambda_1(x), \dots, \lambda_m(x)) : L \rightarrow \mathbf{R}^m$$

is a one-to-one mapping of L onto \mathbf{R}^m which is linear, i.e. for every $i = 1, \dots, m$ one has

$$\begin{aligned} \lambda_i(x+y) &= \lambda_i(x) + \lambda_i(y) \quad \forall (x, y \in L); \\ \lambda_i(\nu x) &= \nu \lambda_i(x) \quad \forall (x \in L, \nu \in \mathbf{R}). \end{aligned} \tag{A.1.2}$$

The reals $\lambda_i(x)$, $i = 1, \dots, m$, are called the coordinates of $x \in L$ in the basis f^1, \dots, f^m .

E.g., the coordinates of a vector $x \in \mathbf{R}^n$ in the standard basis e_1, \dots, e_n of \mathbf{R}^n – the one comprised of the standard basic orths – are exactly the entries of x .

(v) [Dimension formula] *Let L_1, L_2 be linear subspaces of \mathbf{R}^n . Then*

$$\dim(L_1 \cap L_2) + \dim(L_1 + L_2) = \dim(L_1) + \dim(L_2).$$

A.1.5 Linear mappings and matrices

A function $\mathcal{A}(x)$ (another name – *mapping*) defined on \mathbf{R}^n and taking values in \mathbf{R}^m is called *linear*, if it preserves linear operations:

$$\mathcal{A}(x+y) = \mathcal{A}(x) + \mathcal{A}(y) \quad \forall(x, y \in \mathbf{R}^n); \quad \mathcal{A}(\lambda x) = \lambda \mathcal{A}(x) \quad \forall(x \in \mathbf{R}^n, \lambda \in \mathbf{R}).$$

It is immediately seen that a linear mapping from \mathbf{R}^n to \mathbf{R}^m can be represented as multiplication by an $m \times n$ matrix:

$$\mathcal{A}(x) = Ax,$$

and this matrix is uniquely defined by the mapping: the columns A_j of A are just the images of the standard basic orths e_j under the mapping \mathcal{A} :

$$A_j = \mathcal{A}(e_j).$$

Linear mappings from \mathbf{R}^n into \mathbf{R}^m can be added to each other:

$$(\mathcal{A} + \mathcal{B})(x) = \mathcal{A}(x) + \mathcal{B}(x)$$

and multiplied by reals:

$$(\lambda \mathcal{A})(x) = \lambda \mathcal{A}(x),$$

and the results of these operations again are linear mappings from \mathbf{R}^n to \mathbf{R}^m . The addition of linear mappings and multiplication of these mappings by reals correspond to the same operations with the matrices representing the mappings: adding/multiplying by reals mappings, we add, respectively, multiply by reals the corresponding matrices.

Given two linear mappings $\mathcal{A}(x) : \mathbf{R}^n \rightarrow \mathbf{R}^m$ and $\mathcal{B}(y) : \mathbf{R}^m \rightarrow \mathbf{R}^k$, we can build their superposition

$$\mathcal{C}(x) \equiv \mathcal{B}(\mathcal{A}(x)) : \mathbf{R}^n \rightarrow \mathbf{R}^k,$$

which is again a linear mapping, now from \mathbf{R}^n to \mathbf{R}^k . In the language of matrices representing the mappings, the superposition corresponds to matrix multiplication: the $k \times n$ matrix C representing the mapping \mathcal{C} is the product of the matrices representing \mathcal{A} and \mathcal{B} :

$$\mathcal{A}(x) = Ax, \quad \mathcal{B}(y) = By \Rightarrow \mathcal{C}(x) \equiv \mathcal{B}(\mathcal{A}(x)) = B \cdot (Ax) = (BA)x.$$

Important convention. When speaking about adding n -dimensional vectors and multiplying them by reals, it is absolutely unimportant whether we treat the vectors as the column ones, or the row ones, or write down the entries in rectangular tables, or something else. However, when matrix operations (matrix-vector multiplication, transposition, etc.) become involved, it is important whether we treat our vectors as columns, as rows, or as something else. For the sake of definiteness, *from now on we treat all vectors as column ones*, independently of how we refer to them in the text. For example, when saying for the first time what a vector is, we wrote $x = (x_1, \dots, x_n)$, which might suggest that we were speaking about row vectors. We stress that it is not the case, and the only reason for using the notation $x = (x_1, \dots, x_n)$ instead of

the “correct” one $x = \begin{bmatrix} x_1 \\ \vdots \\ x_n \end{bmatrix}$ is to save space and to avoid ugly formulas like $f\left(\begin{bmatrix} x_1 \\ \vdots \\ x_n \end{bmatrix}\right)$ when

speaking about functions with vector arguments. After we have agreed that *there is no such thing as a row vector in this Lecture course*, we can use (and do use) without any harm whatever notation we want.

Exercise A.1 1. Mark in the list below those subsets of \mathbf{R}^n which are linear subspaces, find out their dimensions and point out their bases:

- (a) \mathbf{R}^n
- (b) $\{0\}$
- (c) \emptyset
- (d) $\{x \in \mathbf{R}^n : \sum_{i=1}^n ix_i = 0\}$
- (e) $\{x \in \mathbf{R}^n : \sum_{i=1}^n ix_i^2 = 0\}$
- (f) $\{x \in \mathbf{R}^n : \sum_{i=1}^n ix_i = 1\}$
- (g) $\{x \in \mathbf{R}^n : \sum_{i=1}^n ix_i^2 = 1\}$

2. It is known that L is a subspace of \mathbf{R}^n with exactly one basis. What is L ?
3. Consider the space $\mathbf{R}^{m \times n}$ of $m \times n$ matrices with real entries. As far as linear operations – addition of matrices and multiplication of matrices by reals – are concerned, this space can be treated as certain \mathbf{R}^N .
 - (a) Find the dimension of $\mathbf{R}^{m \times n}$ and point out a basis in this space
 - (b) In the space $\mathbf{R}^{n \times n}$ of square $n \times n$ matrices, there are two interesting subsets: the set \mathbf{S}^n of symmetric matrices $\{A = [A_{ij}] : A_{ij} = A_{ji}\}$ and the set \mathbf{J}^n of skew-symmetric matrices $\{A = [A_{ij}] : A_{ij} = -A_{ji}\}$.
 - i. Verify that both \mathbf{S}^n and \mathbf{J}^n are linear subspaces of $\mathbf{R}^{n \times n}$
 - ii. Find the dimension and point out a basis in \mathbf{S}^n
 - iii. Find the dimension and point out a basis in \mathbf{J}^n
 - iv. What is the sum of \mathbf{S}^n and \mathbf{J}^n ? What is the intersection of \mathbf{S}^n and \mathbf{J}^n ?

A.2 Space \mathbf{R}^n : Euclidean structure

So far, we were interested solely in the algebraic structure of \mathbf{R}^n , or, which is the same, in the properties of the *linear* operations (addition of vectors and multiplication of vectors by scalars) the space is endowed with. Now let us consider another structure on \mathbf{R}^n – the *standard Euclidean structure* – which allows to speak about distances, angles, convergence, etc., and thus makes the space \mathbf{R}^n a much richer mathematical entity.

A.2.1 Euclidean structure

The standard Euclidean structure on \mathbf{R}^n is given by the *standard inner product* – an operation which takes on input two vectors x, y and produces from them a real, specifically, the real

$$\langle x, y \rangle \equiv x^T y = \sum_{i=1}^n x_i y_i$$

The basic properties of the inner product are as follows:

1. [bi-linearity]: The real-valued function $\langle x, y \rangle$ of two vector arguments $x, y \in \mathbf{R}^n$ is linear with respect to every one of the arguments, the other argument being fixed:

$$\begin{aligned}\langle \lambda u + \mu v, y \rangle &= \lambda \langle u, y \rangle + \mu \langle v, y \rangle \quad \forall (u, v, y \in \mathbf{R}^n, \lambda, \mu \in \mathbf{R}) \\ \langle x, \lambda u + \mu v \rangle &= \lambda \langle x, u \rangle + \mu \langle x, v \rangle \quad \forall (x, u, v \in \mathbf{R}^n, \lambda, \mu \in \mathbf{R})\end{aligned}$$

2. [symmetry]: The function $\langle x, y \rangle$ is symmetric:

$$\langle x, y \rangle = \langle y, x \rangle \quad \forall (x, y \in \mathbf{R}^n).$$

3. [positive definiteness]: The quantity $\langle x, x \rangle$ always is nonnegative, and it is zero if and only if x is zero.

Remark A.2.1 The outlined 3 properties – bi-linearity, symmetry and positive definiteness – form a definition of an Euclidean inner product, and there are infinitely many different from each other ways to satisfy these properties; in other words, there are infinitely many different Euclidean inner products on \mathbf{R}^n . The standard inner product $\langle x, y \rangle = x^T y$ is just a particular case of this general notion. Although in the sequel we normally work with the standard inner product, the reader should remember that the facts we are about to recall are valid for all Euclidean inner products, and not only for the standard one.

The notion of an inner product underlies a number of purely algebraic constructions, in particular, those of *inner product representation of linear forms* and of *orthogonal complement*.

A.2.2 Inner product representation of linear forms on \mathbf{R}^n

A *linear form* on \mathbf{R}^n is a real-valued function $f(x)$ on \mathbf{R}^n which is additive ($f(x+y) = f(x)+f(y)$) and homogeneous ($f(\lambda x) = \lambda f(x)$)

Example of linear form: $f(x) = \sum_{i=1}^n ix_i$

Examples of non-linear functions: (1) $f(x) = x_1 + 1$; (2) $f(x) = x_1^2 - x_2^2$; (3) $f(x) = \sin(x_1)$.

When adding/multiplying by reals linear forms, we again get linear forms (scientifically speaking: “linear forms on \mathbf{R}^n form a linear space”). *Euclidean structure allows to identify linear forms on \mathbf{R}^n with vectors from \mathbf{R}^n :*

Theorem A.2.1 Let $\langle \cdot, \cdot \rangle$ be a Euclidean inner product on \mathbf{R}^n .

(i) Let $f(x)$ be a linear form on \mathbf{R}^n . Then there exists a uniquely defined vector $f \in \mathbf{R}^n$ such that the form is just the inner product with f :

$$f(x) = \langle f, x \rangle \quad \forall x$$

(ii) Vice versa, every vector $f \in \mathbf{R}^n$ defines, via the formula

$$f(x) \equiv \langle f, x \rangle,$$

a linear form on \mathbf{R}^n ;

(iii) The above one-to-one correspondence between the linear forms and vectors on \mathbf{R}^n is linear: adding linear forms (or multiplying a linear form by a real), we add (respectively, multiply by the real) the vector(s) representing the form(s).

A.2.3 Orthogonal complement

An Euclidean structure allows to associate with a linear subspace $L \subset \mathbf{R}^n$ another linear subspace L^\perp – the *orthogonal complement* (or the *annulator*) of L ; by definition, L^\perp consists of all vectors which are orthogonal to every vector from L :

$$L^\perp = \{f : \langle f, x \rangle = 0 \quad \forall x \in L\}.$$

Theorem A.2.2 (i) *Twice taken, orthogonal complement recovers the original subspace: whenever L is a linear subspace of \mathbf{R}^n , one has*

$$(L^\perp)^\perp = L;$$

(ii) *The larger is a linear subspace L , the smaller is its orthogonal complement: if $L_1 \subset L_2$ are linear subspaces of \mathbf{R}^n , then $L_1^\perp \supset L_2^\perp$*

(iii) *The intersection of a subspace and its orthogonal complement is trivial, and the sum of these subspaces is the entire \mathbf{R}^n :*

$$L \cap L^\perp = \{0\}, \quad L + L^\perp = \mathbf{R}^n.$$

Remark A.2.2 From Theorem A.2.2.(iii) and the Dimension formula (Theorem A.1.2.(v)) it follows, first, that for every subspace L in \mathbf{R}^n one has

$$\dim(L) + \dim(L^\perp) = n.$$

Second, every vector $x \in \mathbf{R}^n$ admits a unique decomposition

$$x = x_L + x_{L^\perp}$$

into a sum of two vectors: the first of them, x_L , belongs to L , and the second, x_{L^\perp} , belongs to L^\perp . This decomposition is called the *orthogonal decomposition* of x taken with respect to L, L^\perp ; x_L is called the *orthogonal projection* of x onto L , and x_{L^\perp} – the orthogonal projection of x onto the orthogonal complement of L . Both projections depend on x linearly, for example,

$$(x+y)_L = x_L + y_L, \quad (\lambda x)_L = \lambda x_L.$$

The mapping $x \mapsto x_L$ is called the *orthogonal projector* onto L .

A.2.4 Orthonormal bases

A collection of vectors f^1, \dots, f^m is called *orthonormal* w.r.t. Euclidean inner product $\langle \cdot, \cdot \rangle$, if distinct vector from the collection are orthogonal to each other:

$$i \neq j \Rightarrow \langle f^i, f^j \rangle = 0$$

and inner product of every vector f^i with itself is unit:

$$\langle f^i, f^i \rangle = 1, \quad i = 1, \dots, m.$$

Theorem A.2.3 (i) An orthonormal collection f^1, \dots, f^m always is linearly independent and is therefore a basis of its linear span $L = \text{Lin}(f^1, \dots, f^m)$ (such a basis in a linear subspace is called orthonormal). The coordinates of a vector $x \in L$ w.r.t. an orthonormal basis f^1, \dots, f^m of L are given by explicit formulas:

$$x = \sum_{i=1}^m \lambda_i(x) f^i \Leftrightarrow \lambda_i(x) = \langle x, f^i \rangle.$$

Example of an orthonormal basis in \mathbf{R}^n : The standard basis $\{e_1, \dots, e_n\}$ is orthonormal with respect to the standard inner product $\langle x, y \rangle = x^T y$ on \mathbf{R}^n (but is not orthonormal w.r.t. other Euclidean inner products on \mathbf{R}^n).

Proof of (i): Taking inner product of both sides in the equality

$$x = \sum_j \lambda_j(x) f^j$$

with f^i , we get

$$\begin{aligned} \langle x, f_i \rangle &= \left\langle \sum_j \lambda_j(x) f^j, f^i \right\rangle \\ &= \sum_j \lambda_j(x) \langle f^j, f^i \rangle \quad [\text{bilinearity of inner product}] \\ &= \lambda_i(x) \quad [\text{orthonormality of } \{f^i\}] \end{aligned}$$

Similar computation demonstrates that if 0 is represented as a linear combination of f^i with certain coefficients λ_i , then $\lambda_i = \langle 0, f^i \rangle = 0$, i.e., all the coefficients are zero; this means that an orthonormal system is linearly independent.

(ii) If f^1, \dots, f^m is an orthonormal basis in a linear subspace L , then the inner product of two vectors $x, y \in L$ in the coordinates $\lambda_i(\cdot)$ w.r.t. this basis is given by the standard formula

$$\langle x, y \rangle = \sum_{i=1}^m \lambda_i(x) \lambda_i(y).$$

Proof:

$$\begin{aligned} x &= \sum_i \lambda_i(x) f^i, \quad y = \sum_i \lambda_i(y) f^i \\ \Rightarrow \langle x, y \rangle &= \left\langle \sum_i \lambda_i(x) f^i, \sum_i \lambda_i(y) f^i \right\rangle \\ &= \sum_{i,j} \lambda_i(x) \lambda_j(y) \langle f^i, f^j \rangle \quad [\text{bilinearity of inner product}] \\ &= \sum_i \lambda_i(x) \lambda_i(y) \quad [\text{orthonormality of } \{f^i\}] \end{aligned}$$

(iii) Every linear subspace L of \mathbf{R}^n admits an orthonormal basis; moreover, every orthonormal system f^1, \dots, f^m of vectors from L can be extended to an orthonormal basis in L .

Important corollary: All Euclidean spaces of the same dimension are “the same”. Specifically, if L is an m -dimensional space in a space \mathbf{R}^n equipped with an Euclidean inner product $\langle \cdot, \cdot \rangle$, then there exists a one-to-one mapping $x \mapsto A(x)$ of L onto \mathbf{R}^m such that

- The mapping preserves linear operations:

$$A(x + y) = A(x) + A(y) \quad \forall (x, y \in L); \quad A(\lambda x) = \lambda A(x) \quad \forall (x \in L, \lambda \in \mathbf{R});$$

- The mapping converts the $\langle \cdot, \cdot \rangle$ inner product on L into the standard inner product on \mathbf{R}^m :

$$\langle x, y \rangle = (A(x))^T A(y) \quad \forall x, y \in L.$$

Indeed, by (iii) L admits an orthonormal basis f^1, \dots, f^m ; using (ii), one can immediately check that the mapping

$$x \mapsto A(x) = (\lambda_1(x), \dots, \lambda_m(x))$$

which maps $x \in L$ into the m -dimensional vector comprised of the coordinates of x in the basis f^1, \dots, f^m , meets all the requirements.

Proof of (iii) is given by important by its own right *Gram-Schmidt orthogonalization process* as follows. We start with an arbitrary basis h^1, \dots, h^m in L and step by step convert it into an orthonormal basis f^1, \dots, f^m . At the beginning of a step t of the construction, we already have an orthonormal collection f^1, \dots, f^{t-1} such that $\text{Lin}\{f^1, \dots, f^{t-1}\} = \text{Lin}\{h^1, \dots, h^{t-1}\}$. At a step t we

1. Build the vector

$$g^t = h^t - \sum_{j=1}^{t-1} \langle h^t, f^j \rangle f^j.$$

It is easily seen (check it!) that

- (a) One has

$$\text{Lin}\{f^1, \dots, f^{t-1}, g^t\} = \text{Lin}\{h^1, \dots, h^t\}; \quad (\text{A.2.1})$$

- (b) $g^t \neq 0$ (derive this fact from (A.2.1) and the linear independence of the collection h^1, \dots, h^m);
- (c) g^t is orthogonal to f^1, \dots, f^{t-1}

2. Since $g^t \neq 0$, the quantity $\langle g^t, g^t \rangle$ is positive (positive definiteness of the inner product), so that the vector

$$f^t = \frac{1}{\sqrt{\langle g^t, g^t \rangle}} g^t$$

is well defined. It is immediately seen (check it!) that the collection f^1, \dots, f^t is orthonormal and

$$\text{Lin}\{f^1, \dots, f^t\} = \text{Lin}\{f^1, \dots, f^{t-1}, g^t\} = \text{Lin}\{h^1, \dots, h^t\}.$$

Step t of the orthogonalization process is completed.

After m steps of the optimization process, we end up with an orthonormal system f^1, \dots, f^m of vectors from L such that

$$\text{Lin}\{f^1, \dots, f^m\} = \text{Lin}\{h^1, \dots, h^m\} = L,$$

so that f^1, \dots, f^m is an orthonormal basis in L .

The construction can be easily modified (do it!) to extend a given orthonormal system of vectors from L to an orthonormal basis of L .

- Exercise A.2**
1. What is the orthogonal complement (w.r.t. the standard inner product) of the subspace $\{x \in \mathbf{R}^n : \sum_{i=1}^n x_i = 0\}$ in \mathbf{R}^n ?
 2. Find an orthonormal basis (w.r.t. the standard inner product) in the linear subspace $\{x \in \mathbf{R}^n : x_1 = 0\}$ of \mathbf{R}^n

3. Let L be a linear subspace of \mathbf{R}^n , and f^1, \dots, f^m be an orthonormal basis in L . Prove that for every $x \in \mathbf{R}^n$, the orthoprojection x_L of x onto L is given by the formula

$$x_L = \sum_{i=1}^m (x^T f^i) f^i.$$

4. Let L_1, L_2 be linear subspaces in \mathbf{R}^n . Verify the formulas

$$(L_1 + L_2)^\perp = L_1^\perp \cap L_2^\perp; \quad (L_1 \cap L_2)^\perp = L_1^\perp + L_2^\perp.$$

5. Consider the space of $m \times n$ matrices $\mathbf{R}^{m \times n}$, and let us equip it with the “standard inner product” (called the Frobenius inner product)

$$\langle A, B \rangle = \sum_{i,j} A_{ij} B_{ij}$$

(as if we were treating $m \times n$ matrices as mn -dimensional vectors, writing the entries of the matrices column by column, and then taking the standard inner product of the resulting long vectors).

- (a) Verify that in terms of matrix multiplication the Frobenius inner product can be written as

$$\langle A, B \rangle = \text{Tr}(AB^T),$$

where $\text{Tr}(C)$ is the trace (the sum of diagonal elements) of a square matrix C .

- (b) Build an orthonormal basis in the linear subspace \mathbf{S}^n of symmetric $n \times n$ matrices
(c) What is the orthogonal complement of the subspace \mathbf{S}^n of symmetric $n \times n$ matrices in the space $\mathbf{R}^{n \times n}$ of square $n \times n$ matrices?

- (d) Find the orthogonal decomposition, w.r.t. \mathbf{S}^2 , of the matrix $\begin{bmatrix} 1 & 2 \\ 3 & 4 \end{bmatrix}$

A.3 Affine subspaces in \mathbf{R}^n

Many of events to come will take place not in the entire \mathbf{R}^n , but in its *affine subspaces* which, geometrically, are planes of different dimensions in \mathbf{R}^n . Let us become acquainted with these subspaces.

A.3.1 Affine subspaces and affine hulls

Definition of an affine subspace. Geometrically, a linear subspace L of \mathbf{R}^n is a special plane – the one passing through the origin of the space (i.e., containing the zero vector). To get an arbitrary plane M , it suffices to subject an appropriate special plane L to a translation – to add to all points from L a fixed *shifting vector* a . This geometric intuition leads to the following

Definition A.3.1 [Affine subspace] An affine subspace (a plane) in \mathbf{R}^n is a set of the form

$$M = a + L = \{y = a + x \mid x \in L\}, \tag{A.3.1}$$

where L is a linear subspace in \mathbf{R}^n and a is a vector from \mathbf{R}^{n-2} .

E.g., shifting the linear subspace L comprised of vectors with zero first entry by a vector $a = (a_1, \dots, a_n)$, we get the set $M = a + L$ of all vectors x with $x_1 = a_1$; according to our terminology, this is an affine subspace.

Immediate question about the notion of an affine subspace is: what are the “degrees of freedom” in decomposition (A.3.1) – how “strict” M determines a and L ? The answer is as follows:

Proposition A.3.1 *The linear subspace L in decomposition (A.3.1) is uniquely defined by M and is the set of all differences of the vectors from M :*

$$L = M - M = \{x - y \mid x, y \in M\}. \quad (\text{A.3.2})$$

The shifting vector a is not uniquely defined by M and can be chosen as an arbitrary vector from M .

A.3.2 Intersections of affine subspaces, affine combinations and affine hulls

An immediate conclusion of Proposition A.3.1 is as follows:

Corollary A.3.1 *Let $\{M_\alpha\}$ be an arbitrary family of affine subspaces in \mathbf{R}^n , and assume that the set $M = \cap_\alpha M_\alpha$ is nonempty. Then M is an affine subspace.*

From Corollary A.3.1 it immediately follows that for every nonempty subset Y of \mathbf{R}^n there exists the smallest affine subspace containing Y – the intersection of all affine subspaces containing Y . This smallest affine subspace containing Y is called the *affine hull* of Y (notation: $\text{Aff}(Y)$).

All this resembles a lot the story about linear spans. Can we further extend this analogy and to get a description of the affine hull $\text{Aff}(Y)$ in terms of elements of Y similar to the one of the linear span (“linear span of X is the set of all linear combinations of vectors from X ”)? Sure we can!

Let us choose somehow a point $y_0 \in Y$, and consider the set

$$X = Y - y_0.$$

All affine subspaces containing Y should contain also y_0 and therefore, by Proposition A.3.1, can be represented as $M = y_0 + L$, L being a linear subspace. It is absolutely evident that an affine subspace $M = y_0 + L$ contains Y if and only if the subspace L contains X , and that the larger is L , the larger is M :

$$L \subset L' \Rightarrow M = y_0 + L \subset M' = y_0 + L'.$$

Thus, to find the smallest among *affine subspaces containing Y* , it suffices to find the smallest among the *linear subspaces containing X* and to translate the latter space by y_0 :

$$\text{Aff}(Y) = y_0 + \text{Lin}(X) = y_0 + \text{Lin}(Y - y_0). \quad (\text{A.3.3})$$

²⁾according to our convention on arithmetic of sets, I was supposed to write in (A.3.1) $\{a\} + L$ instead of $a + L$ – we did not define arithmetic sum of a vector and a set. Usually people ignore this difference and omit the brackets when writing down singleton sets in similar expressions: we shall write $a + L$ instead of $\{a\} + L$, $\mathbf{R}d$ instead of $\mathbf{R}\{d\}$, etc.

Now, we know what is $\text{Lin}(Y - y_0)$ – this is a set of all linear combinations of vectors from $Y - y_0$, so that a generic element of $\text{Lin}(Y - y_0)$ is

$$x = \sum_{i=1}^k \mu_i(y_i - y_0) \quad [k \text{ may depend of } x]$$

with $y_i \in Y$ and real coefficients μ_i . It follows that the generic element of $\text{Aff}(Y)$ is

$$y = y_0 + \sum_{i=1}^k \mu_i(y_i - y_0) = \sum_{i=0}^k \lambda_i y_i,$$

where

$$\lambda_0 = 1 - \sum_i \mu_i, \quad \lambda_i = \mu_i, \quad i \geq 1.$$

We see that a generic element of $\text{Aff}(Y)$ is a linear combination of vectors from Y . Note, however, that the coefficients λ_i in this combination are not completely arbitrary: their sum is equal to 1. Linear combinations of this type – with the unit sum of coefficients – have a special name – they are called affine combinations.

We have seen that every vector from $\text{Aff}(Y)$ is an affine combination of vectors of Y . Whether the inverse is true, i.e., whether $\text{Aff}(Y)$ contains all affine combinations of vectors from Y ? The answer is positive. Indeed, if

$$y = \sum_{i=1}^k \lambda_i y_i$$

is an affine combination of vectors from Y , then, using the equality $\sum_i \lambda_i = 1$, we can write it also as

$$y = y_0 + \sum_{i=1}^k \lambda_i(y_i - y_0),$$

y_0 being the “marked” vector we used in our previous reasoning, and the vector of this form, as we already know, belongs to $\text{Aff}(Y)$. Thus, we come to the following

Proposition A.3.2 [Structure of affine hull]

$$\text{Aff}(Y) = \{\text{the set of all affine combinations of vectors from } Y\}.$$

When Y itself is an affine subspace, it, of course, coincides with its affine hull, and the above Proposition leads to the following

Corollary A.3.2 *An affine subspace M is closed with respect to taking affine combinations of its members – every combination of this type is a vector from M . Vice versa, a nonempty set which is closed with respect to taking affine combinations of its members is an affine subspace.*

A.3.3 Affinely spanning sets, affinely independent sets, affine dimension

Affine subspaces are closely related to linear subspaces, and the basic notions associated with linear subspaces have natural and useful affine analogies. Here we introduce these notions and discuss their basic properties.

Affinely spanning sets. Let $M = a + L$ be an affine subspace. We say that a subset Y of M is *affinely spanning* for M (we say also that Y spans M affinely, or that M is affinely spanned by Y), if $M = \text{Aff}(Y)$, or, which is the same due to Proposition A.3.2, if every point of M is an affine combination of points from Y . An immediate consequence of the reasoning of the previous Section is as follows:

Proposition A.3.3 *Let $M = a + L$ be an affine subspace and Y be a subset of M , and let $y_0 \in Y$. The set Y affinely spans $M - M = \text{Aff}(Y) - y_0$ if and only if the set*

$$X = Y - y_0$$

spans the linear subspace L : $L = \text{Lin}(X)$.

Affinely independent sets. A linearly independent set x_1, \dots, x_k is a set such that no nontrivial linear combination of x_1, \dots, x_k equals to zero. An equivalent definition is given by Theorem A.1.2.(iv): x_1, \dots, x_k are linearly independent, if the coefficients in a linear combination

$$x = \sum_{i=1}^k \lambda_i x_i$$

are uniquely defined by the value x of the combination. This equivalent form reflects the essence of the matter – what we indeed need, is the uniqueness of the coefficients in expansions. Accordingly, this equivalent form is the prototype for the notion of an affinely independent set: we want to introduce this notion in such a way that the coefficients λ_i in an *affine* combination

$$y = \sum_{i=0}^k \lambda_i y_i$$

of “affinely independent” set of vectors y_0, \dots, y_k would be uniquely defined by y . Non-uniqueness would mean that

$$\sum_{i=0}^k \lambda_i y_i = \sum_{i=0}^k \lambda'_i y_i$$

for two different collections of coefficients λ_i and λ'_i with unit sums of coefficients; if it is the case, then

$$\sum_{i=0}^m (\lambda_i - \lambda'_i) y_i = 0,$$

so that y_i 's are linearly dependent and, moreover, there exists a nontrivial zero combination of them with zero sum of coefficients (since $\sum_i (\lambda_i - \lambda'_i) = \sum_i \lambda_i - \sum_i \lambda'_i = 1 - 1 = 0$). Our reasoning can be inverted – if there exists a nontrivial linear combination of y_i 's with zero sum of coefficients which is zero, then the coefficients in the representation of a vector as an affine combination of y_i 's are not uniquely defined. Thus, in order to get uniqueness we should for sure forbid relations

$$\sum_{i=0}^k \mu_i y_i = 0$$

with nontrivial zero sum coefficients μ_i . Thus, we have motivated the following

Definition A.3.2 [Affinely independent set] A collection y_0, \dots, y_k of n -dimensional vectors is called affinely independent, if no nontrivial linear combination of the vectors with zero sum of coefficients is zero:

$$\sum_{i=1}^k \lambda_i y_i = 0, \quad \sum_{i=0}^k \lambda_i = 0 \Rightarrow \lambda_0 = \lambda_1 = \dots = \lambda_k = 0.$$

With this definition, we get the result completely similar to the one of Theorem A.1.2.(iv):

Corollary A.3.3 Let y_0, \dots, y_k be affinely independent. Then the coefficients λ_i in an affine combination

$$y = \sum_{i=0}^k \lambda_i y_i \quad [\sum_i \lambda_i = 1]$$

of the vectors y_0, \dots, y_k are uniquely defined by the value y of the combination.

Verification of affine independence of a collection can be immediately reduced to verification of linear independence of closely related collection:

Proposition A.3.4 $k+1$ vectors y_0, \dots, y_k are affinely independent if and only if the k vectors $(y_1 - y_0), (y_2 - y_0), \dots, (y_k - y_0)$ are linearly independent.

From the latter Proposition it follows, e.g., that the collection $0, e_1, \dots, e_n$ comprised of the origin and the standard basic orths is affinely independent. Note that this collection is linearly dependent (as every collection containing zero). You should definitely know the difference between the two notions of independence we deal with: linear independence means that no nontrivial linear combination of the vectors can be zero, while affine independence means that no nontrivial linear combination from certain restricted class of them (with zero sum of coefficients) can be zero. Therefore, there are more affinely independent sets than the linearly independent ones: a linearly independent set is for sure affinely independent, but not vice versa.

Affine bases and affine dimension. Propositions A.3.2 and A.3.3 reduce the notions of affine spanning/affine independent sets to the notions of spanning/linearly independent ones. Combined with Theorem A.1.2, they result in the following analogies of the latter two statements:

Proposition A.3.5 [Affine dimension] Let $M = a + L$ be an affine subspace in \mathbf{R}^n . Then the following two quantities are finite integers which are equal to each other:

- (i) minimal # of elements in the subsets of M which affinely span M ;
- (ii) maximal # of elements in affine independent subsets of M .

The common value of these two integers is by 1 more than the dimension $\dim L$ of L .

By definition, the affine dimension of an affine subspace $M = a + L$ is the dimension $\dim L$ of L . Thus, if M is of affine dimension k , then the minimal cardinality of sets affinely spanning M , same as the maximal cardinality of affine independent subsets of M , is $k+1$.

Theorem A.3.1 [Affine bases] Let $M = a + L$ be an affine subspace in \mathbf{R}^n .

A. Let $Y \subset M$. The following three properties of X are equivalent:

- (i) Y is an affine independent set which affinely spans M ;
- (ii) Y is affine independent and contains $1 + \dim L$ elements;

(iii) Y affinely spans M and contains $1 + \dim L$ elements.

A subset Y of M possessing the indicated equivalent to each other properties is called an affine basis of M . Affine bases in M are exactly the collections $y_0, \dots, y_{\dim L}$ such that $y_0 \in M$ and $(y_1 - y_0), \dots, (y_{\dim L} - y_0)$ is a basis in L .

B. Every affinely independent collection of vectors of M either itself is an affine basis of M , or can be extended to such a basis by adding new vectors. In particular, there exists affine basis of M .

C. Given a set Y which affinely spans M , you can always extract from this set an affine basis of M .

We already know that the standard basic orths e_1, \dots, e_n form a basis of the entire space \mathbf{R}^n . And what about affine bases in \mathbf{R}^n ? According to Theorem A.3.1.A, you can choose as such a basis a collection $e_0, e_0 + e_1, \dots, e_0 + e_n$, e_0 being an arbitrary vector.

Barycentric coordinates. Let M be an affine subspace, and let y_0, \dots, y_k be an affine basis of M . Since the basis, by definition, affinely spans M , every vector y from M is an affine combination of the vectors of the basis:

$$y = \sum_{i=0}^k \lambda_i y_i \quad [\sum_{i=0}^k \lambda_i = 1],$$

and since the vectors of the affine basis are affinely independent, the coefficients of this combination are uniquely defined by y (Corollary A.3.3). These coefficients are called *barycentric coordinates* of y with respect to the affine basis in question. In contrast to the usual coordinates with respect to a (linear) basis, the barycentric coordinates could not be quite arbitrary: their sum should be equal to 1.

A.3.4 Dual description of linear subspaces and affine subspaces

To the moment we have introduced the notions of linear subspace and affine subspace and have presented a scheme of generating these entities: to get, e.g., a linear subspace, you start from an arbitrary nonempty set $X \subset \mathbf{R}^n$ and add to it all linear combinations of the vectors from X . When replacing linear combinations with the affine ones, you get a way to generate affine subspaces.

The just indicated way of generating linear subspaces/affine subspaces resembles the approach of a worker building a house: he starts with the base and then adds to it new elements until the house is ready. There exists, anyhow, an approach of an artist creating a sculpture: he takes something large and then deletes extra parts of it. Is there something like “artist’s way” to represent linear subspaces and affine subspaces? The answer is positive and very instructive.

A.3.4.1 Affine subspaces and systems of linear equations

Let L be a linear subspace. According to Theorem A.2.2.(i), it is an orthogonal complement – namely, the orthogonal complement to the linear subspace L^\perp . Now let a_1, \dots, a_m be a finite spanning set in L^\perp . A vector x which is orthogonal to a_1, \dots, a_m is orthogonal to the entire L^\perp (since every vector from L^\perp is a linear combination of a_1, \dots, a_m and the inner product is bilinear); and of course vice versa, a vector orthogonal to the entire L^\perp is orthogonal to a_1, \dots, a_m . We see that

$$L = (L^\perp)^\perp = \{x \mid a_i^T x = 0, i = 1, \dots, m\}. \tag{A.3.4}$$

Thus, we get a very important, although simple,

Proposition A.3.6 [“Outer” description of a linear subspace] *Every linear subspace L in \mathbf{R}^n is a set of solutions to a homogeneous linear system of equations*

$$a_i^T x = 0, \quad i = 1, \dots, m, \quad (\text{A.3.5})$$

given by properly chosen m and vectors a_1, \dots, a_m .

Proposition A.3.6 is an “if and only if” statement: as we remember from Example A.1.3.A.4, solution set to a homogeneous system of linear equations with n variables always is a linear subspace in \mathbf{R}^n .

From Proposition A.3.6 and the facts we know about the dimension we can easily derive several important consequences:

- Systems (A.3.5) which define a given linear subspace L are exactly the systems given by the vectors a_1, \dots, a_m which span L^\perp ³⁾
- The smallest possible number m of equations in (A.3.5) is the dimension of L^\perp , i.e., by Remark A.2.2, is $\text{codim } L \equiv n - \dim L$ ⁴⁾

Now, an affine subspace M is, by definition, a translation of a linear subspace: $M = a + L$. As we know, vectors x from L are exactly the solutions of certain *homogeneous* system of linear equations

$$a_i^T x = 0, \quad i = 1, \dots, m.$$

It is absolutely clear that adding to these vectors a fixed vector a , we get exactly the set of solutions to the *inhomogeneous* solvable linear system

$$a_i^T x = b_i \equiv a_i^T a, \quad i = 1, \dots, m.$$

Vice versa, the set of solutions to a *solvable* system of linear equations

$$a_i^T x = b_i, \quad i = 1, \dots, m,$$

with n variables is the sum of a particular solution to the system and the solution set to the corresponding homogeneous system (the latter set, as we already know, is a linear subspace in \mathbf{R}^n), i.e., is an affine subspace. Thus, we get the following

Proposition A.3.7 [“Outer” description of an affine subspace]

Every affine subspace $M = a + L$ in \mathbf{R}^n is a set of solutions to a solvable linear system of equations

$$a_i^T x = b_i, \quad i = 1, \dots, m, \quad (\text{A.3.6})$$

given by properly chosen m and vectors a_1, \dots, a_m .

Vice versa, the set of all solutions to a solvable system of linear equations with n variables is an affine subspace in \mathbf{R}^n .

The linear subspace L associated with M is exactly the set of solutions of the homogeneous (with the right hand side set to 0) version of system (A.3.6).

We see, in particular, that an affine subspace always is closed.

³⁾the reasoning which led us to Proposition A.3.6 says that $[a_1, \dots, a_m \text{ span } L^\perp] \Rightarrow [(\text{A.3.5}) \text{ defines } L]$; now we claim that the inverse also is true

⁴⁾to make this statement true also in the extreme case when $L = \mathbf{R}^n$ (i.e., when $\text{codim } L = 0$), we from now on make a convention that an *empty* set of equations or inequalities defines, as the solution set, the entire space

Comment. The “outer” description of a linear subspace/affine subspace – the “artist’s” one – is in many cases much more useful than the “inner” description via linear/affine combinations (the “worker’s” one). E.g., with the outer description it is very easy to check whether a given vector belongs or does not belong to a given linear subspace/affine subspace, which is not that easy with the inner one⁵⁾. In fact both descriptions are “complementary” to each other and perfectly well work in parallel: what is difficult to see with one of them, is clear with another. The idea of using “inner” and “outer” descriptions of the entities we meet with – linear subspaces, affine subspaces, convex sets, optimization problems – the general idea of *duality* – is, I would say, the main driving force of Convex Analysis and Optimization, and in the sequel we would all the time meet with different implementations of this fundamental idea.

A.3.5 Structure of the simplest affine subspaces

This small subsection deals mainly with terminology. According to their dimension, affine subspaces in \mathbf{R}^n are named as follows:

- Subspaces of dimension 0 are translations of the only 0-dimensional linear subspace $\{0\}$, i.e., are singleton sets – vectors from \mathbf{R}^n . These subspaces are called *points*; a point is a solution to a square system of linear equations with nonsingular matrix.
- Subspaces of dimension 1 (lines). These subspaces are translations of one-dimensional linear subspaces of \mathbf{R}^n . A one-dimensional linear subspace has a single-element basis given by a nonzero vector d and is comprised of all multiples of this vector. Consequently, line is a set of the form

$$\{y = a + td \mid t \in \mathbf{R}\}$$

given by a pair of vectors a (the origin of the line) and d (the direction of the line), $d \neq 0$. The origin of the line and its direction are not uniquely defined by the line; you can choose as origin any point on the line and multiply a particular direction by nonzero reals.

In the barycentric coordinates a line is described as follows:

$$l = \{\lambda_0 y_0 + \lambda_1 y_1 \mid \lambda_0 + \lambda_1 = 1\} = \{\lambda y_0 + (1 - \lambda) y_1 \mid \lambda \in \mathbf{R}\},$$

where y_0, y_1 is an affine basis of l ; you can choose as such a basis any pair of distinct points on the line.

The “outer” description a line is as follows: it is the set of solutions to a linear system with n variables and $n - 1$ linearly independent equations.

- Subspaces of dimension > 2 and $< n - 1$ have no special names; sometimes they are called affine planes of such and such dimension.
- Affine subspaces of dimension $n - 1$, due to important role they play in Convex Analysis, have a special name – they are called *hyperplanes*. The outer description of a hyperplane is that a hyperplane is the solution set of a *single* linear equation

$$a^T x = b$$

⁵⁾in principle it is not difficult to certify that a given point belongs to, say, a linear subspace given as the linear span of some set – it suffices to point out a representation of the point as a linear combination of vectors from the set. But how could you certify that the point does *not* belong to the subspace?

with nontrivial left hand side ($a \neq 0$). In other words, a hyperplane is the level set $a(x) = \text{const}$ of a nonconstant linear form $a(x) = a^T x$.

- The “largest possible” affine subspace – the one of dimension n – is unique and is the entire \mathbf{R}^n . This subspace is given by an empty system of linear equations.

A.4 Space \mathbf{R}^n : metric structure and topology

Euclidean structure on the space \mathbf{R}^n gives rise to a number of extremely important *metric* notions – distances, convergence, etc. For the sake of definiteness, we associate these notions with the standard inner product $x^T y$.

A.4.1 Euclidean norm and distances

By positive definiteness, the quantity $x^T x$ always is nonnegative, so that the quantity

$$|x| \equiv \|x\|_2 = \sqrt{x^T x} = \sqrt{x_1^2 + x_2^2 + \dots + x_n^2}$$

is well-defined; this quantity is called the (standard) *Euclidean norm* of vector x (or simply the norm of x) and is treated as the distance from the origin to x . The distance between two arbitrary points $x, y \in \mathbf{R}^n$ is, by definition, the norm $|x - y|$ of the difference $x - y$. The notions we have defined satisfy all basic requirements on the general notions of a norm and distance, specifically:

1. Positivity of norm: The norm of a vector always is nonnegative; it is zero if and only if the vector is zero:

$$|x| \geq 0 \quad \forall x; \quad |x| = 0 \Leftrightarrow x = 0.$$

2. Homogeneity of norm: When a vector is multiplied by a real, its norm is multiplied by the absolute value of the real:

$$|\lambda x| = |\lambda| \cdot |x| \quad \forall (x \in \mathbf{R}^n, \lambda \in \mathbf{R}).$$

3. Triangle inequality: Norm of the sum of two vectors is \leq the sum of their norms:

$$|x + y| \leq |x| + |y| \quad \forall (x, y \in \mathbf{R}^n).$$

In contrast to the properties of positivity and homogeneity, which are absolutely evident, the Triangle inequality is not trivial and definitely requires a proof. The proof goes through a fact which is extremely important by its own right – the *Cauchy Inequality*, which perhaps is the most frequently used inequality in Mathematics:

Theorem A.4.1 [Cauchy's Inequality] *The absolute value of the inner product of two vectors does not exceed the product of their norms:*

$$|x^T y| \leq |x||y| \quad \forall (x, y \in \mathbf{R}^n)$$

and is equal to the product of the norms if and only if one of the vectors is proportional to the other one:

$$|x^T y| = |x||y| \Leftrightarrow \{\exists \alpha : x = \alpha y \text{ or } \exists \beta : y = \beta x\}$$

Proof is immediate: we may assume that both x and y are nonzero (otherwise the Cauchy inequality clearly is equality, and one of the vectors is constant times (specifically, zero times) the other one, as announced in Theorem). Assuming $x, y \neq 0$, consider the function

$$f(\lambda) = (x - \lambda y)^T(x - \lambda y) = x^T x - 2\lambda x^T y + \lambda^2 y^T y.$$

By positive definiteness of the inner product, this function – which is a second order polynomial – is nonnegative on the entire axis, whence the discriminant of the polynomial

$$(x^T y)^2 - (x^T x)(y^T y)$$

is nonpositive:

$$(x^T y)^2 \leq (x^T x)(y^T y).$$

Taking square roots of both sides, we arrive at the Cauchy Inequality. We also see that the inequality is equality if and only if the discriminant of the second order polynomial $f(\lambda)$ is zero, i.e., if and only if the polynomial has a (multiple) real root; but due to positive definiteness of inner product, $f(\cdot)$ has a root λ if and only if $x = \lambda y$, which proves the second part of Theorem. ■

From Cauchy's Inequality to the Triangle Inequality: Let $x, y \in \mathbf{R}^n$. Then

$$\begin{aligned} |x + y|^2 &= (x + y)^T(x + y) && [\text{definition of norm}] \\ &= x^T x + y^T y + 2x^T y && [\text{opening parentheses}] \\ &\leq \underbrace{x^T x}_{|x|^2} + \underbrace{y^T y}_{|y|^2} + 2|x||y| && [\text{Cauchy's Inequality}] \\ &= (|x| + |y|)^2 \\ \Rightarrow |x + y| &\leq |x| + |y| \end{aligned}$$

■

The properties of norm (i.e., of the distance to the origin) we have established induce properties of the distances between pairs of arbitrary points in \mathbf{R}^n , specifically:

1. Positivity of distances: The distance $|x - y|$ between two points is positive, except for the case when the points coincide ($x = y$), when the distance between x and y is zero;
2. Symmetry of distances: The distance from x to y is the same as the distance from y to x :

$$|x - y| = |y - x|;$$

3. Triangle inequality for distances: For every three points x, y, z , the distance from x to z does not exceed the sum of distances between x and y and between y and z :

$$|z - x| \leq |y - x| + |z - y| \quad \forall (x, y, z \in \mathbf{R}^n)$$

A.4.2 Convergence

Equipped with distances, we can define the fundamental notion of convergence of a sequence of vectors. Specifically, we say that a sequence x^1, x^2, \dots of vectors from \mathbf{R}^n converges to a vector \bar{x} , or, equivalently, that \bar{x} is the limit of the sequence $\{x^i\}$ (notation: $\bar{x} = \lim_{i \rightarrow \infty} x^i$), if the distances from \bar{x} to x^i go to 0 as $i \rightarrow \infty$:

$$\bar{x} = \lim_{i \rightarrow \infty} x^i \Leftrightarrow |\bar{x} - x^i| \rightarrow 0, i \rightarrow \infty,$$

or, which is the same, for every $\epsilon > 0$ there exists $i = i(\epsilon)$ such that the distance between every point x^i , $i \geq i(\epsilon)$, and \bar{x} does not exceed ϵ :

$$\left\{ |\bar{x} - x^i| \rightarrow 0, i \rightarrow \infty \right\} \Leftrightarrow \left\{ \forall \epsilon > 0 \exists i(\epsilon) : i \geq i(\epsilon) \Rightarrow |\bar{x} - x^i| \leq \epsilon \right\}.$$

Exercise A.3 Verify the following facts:

1. $\bar{x} = \lim_{i \rightarrow \infty} x^i$ if and only if for every $j = 1, \dots, n$ the coordinates $\# j$ of the vectors x^i converge, as $i \rightarrow \infty$, to the coordinate $\# j$ of the vector \bar{x} ;
2. If a sequence converges, its limit is uniquely defined;
3. Convergence is compatible with linear operations:
 - if $x^i \rightarrow x$ and $y^i \rightarrow y$ as $i \rightarrow \infty$, then $x^i + y^i \rightarrow x + y$ as $i \rightarrow \infty$;
 - if $x^i \rightarrow x$ and $\lambda_i \rightarrow \lambda$ as $i \rightarrow \infty$, then $\lambda_i x^i \rightarrow \lambda x$ as $i \rightarrow \infty$.

A.4.3 Closed and open sets

After we have at our disposal distance and convergence, we can speak about *closed* and *open* sets:

- A set $X \subset \mathbf{R}^n$ is called *closed*, if it contains limits of all converging sequences of elements of X :

$$\left\{ x^i \in X, x = \lim_{i \rightarrow \infty} x^i \right\} \Rightarrow x \in X$$

- A set $X \subset \mathbf{R}^n$ is called *open*, if whenever x belongs to X , all points close enough to x also belong to X :

$$\forall (x \in X) \exists (\delta > 0) : |x' - x| < \delta \Rightarrow x' \in X.$$

An open set containing a point x is called a *neighbourhood* of x .

Examples of closed sets: (1) \mathbf{R}^n ; (2) \emptyset ; (3) the sequence $x^i = (i, 0, \dots, 0)$, $i = 1, 2, 3, \dots$;

(4) $\{x \in \mathbf{R}^n : \sum_{i=1}^n a_{ij} x_j = 0, i = 1, \dots, m\}$ (in other words: a linear subspace in \mathbf{R}^n

always is closed, see Proposition A.3.6); (5) $\{x \in \mathbf{R}^n : \sum_{i=1}^n a_{ij} x_j = b_i, i = 1, \dots, m\}$

(in other words: an affine subset of \mathbf{R}^n always is closed, see Proposition A.3.7); (6) Any finite subset of \mathbf{R}^n

Examples of non-closed sets: (1) $\mathbf{R}^n \setminus \{0\}$; (2) the sequence $x^i = (1/i, 0, \dots, 0)$, $i = 1, 2, 3, \dots$;

(3) $\{x \in \mathbf{R}^n : x_j > 0, j = 1, \dots, n\}$; (4) $\{x \in \mathbf{R}^n : \sum_{i=1}^n x_j > 5\}$.

Examples of open sets: (1) \mathbf{R}^n ; (2) \emptyset ; (3) $\{x \in \mathbf{R}^n : \sum_{j=1}^n a_{ij} x_j > b_j, i = 1, \dots, m\}$; (4)

complement of a finite set.

Examples of non-open sets: (1) A nonempty finite set; (2) the sequence $x^i = (1/i, 0, \dots, 0)$, $i = 1, 2, 3, \dots$, and the sequence $x^i = (i, 0, 0, \dots, 0)$, $i = 1, 2, 3, \dots$;

(3) $\{x \in \mathbf{R}^n : x_j \geq 0, j = 1, \dots, n\}$; (4) $\{x \in \mathbf{R}^n : \sum_{i=1}^n x_j \geq 5\}$.

Exercise A.4 *Mark in the list to follows those sets which are closed and those which are open:*

1. All vectors with integer coordinates
2. All vectors with rational coordinates
3. All vectors with positive coordinates
4. All vectors with nonnegative coordinates
5. $\{x : |x| < 1\};$
6. $\{x : |x| = 1\};$
7. $\{x : |x| \leq 1\};$
8. $\{x : |x| \geq 1\};$
9. $\{x : |x| > 1\};$
10. $\{x : 1 < |x| \leq 2\}.$

Verify the following facts

1. A set $X \subset \mathbf{R}^n$ is closed if and only if its complement $\bar{X} = \mathbf{R}^n \setminus X$ is open;
2. Intersection of every family (finite or infinite) of closed sets is closed. Union of every family (finite or infinite) of open sets is open.
3. Union of finitely many closed sets is closed. Intersection of finitely many open sets is open.

A.4.4 Local compactness of \mathbf{R}^n

A fundamental fact about convergence in \mathbf{R}^n , which in certain sense is characteristic for this series of spaces, is the following

Theorem A.4.2 *From every bounded sequence $\{x^i\}_{i=1}^\infty$ of points from \mathbf{R}^n one can extract a converging subsequence $\{x^{i_j}\}_{j=1}^\infty$. Equivalently: A closed and bounded subset X of \mathbf{R}^n is compact, i.e., a set possessing the following two equivalent to each other properties:*

- (i) *From every sequence of elements of X one can extract a subsequence which converges to certain point of X ;*
- (ii) *From every open covering of X (i.e., a family $\{U_\alpha\}_{\alpha \in A}$ of open sets such that $X \subset \bigcup_{\alpha \in A} U_\alpha$) one can extract a finite sub-covering, i.e., a finite subset of indices $\alpha_1, \dots, \alpha_N$ such that $X \subset \bigcup_{i=1}^N U_{\alpha_i}$.*

A.5 Continuous functions on \mathbf{R}^n

A.5.1 Continuity of a function

Let $X \subset \mathbf{R}^n$ and $f(x) : X \rightarrow \mathbf{R}^m$ be a function (another name – mapping) defined on X and taking values in \mathbf{R}^m .

1. f is called *continuous at a point* $\bar{x} \in X$, if for every sequence x^i of points of X converging to \bar{x} the sequence $f(x^i)$ converges to $f(\bar{x})$. Equivalent definition:

$f : X \rightarrow \mathbf{R}^m$ is continuous at $\bar{x} \in X$, if for every $\epsilon > 0$ there exists $\delta > 0$ such that

$$x \in X, |x - \bar{x}| < \delta \Rightarrow |f(x) - f(\bar{x})| < \epsilon.$$

2. f is called *continuous on X* , if f is continuous at every point from X . Equivalent definition: f preserves convergence: whenever a sequence of points $x^i \in X$ converges to a point $x \in X$, the sequence $f(x^i)$ converges to $f(x)$.

Examples of continuous mappings:

1. An *affine mapping*

$$f(x) = \begin{bmatrix} \sum_{j=1}^m A_{1j}x_j + b_1 \\ \vdots \\ \sum_{j=1}^m A_{mj}x_j + b_m \end{bmatrix} \equiv Ax + b : \mathbf{R}^n \rightarrow \mathbf{R}^m$$

is continuous on the entire \mathbf{R}^n (and thus – on every subset of \mathbf{R}^n) (check it!).

2. The norm $|x|$ is a continuous on \mathbf{R}^n (and thus – on every subset of \mathbf{R}^n) real-valued function (check it!).

Exercise A.5 • Consider the function

$$f(x_1, x_2) = \begin{cases} \frac{x_1^2 - x_2^2}{x_1^2 + x_2^2}, & (x_1, x_2) \neq 0 \\ 0, & x_1 = x_2 = 0 \end{cases} : \mathbf{R}^2 \rightarrow \mathbf{R}.$$

Check whether this function is continuous on the following sets:

1. \mathbf{R}^2 ;
 2. $\mathbf{R}^2 \setminus \{0\}$;
 3. $\{x \in \mathbf{R}^2 : x_1 = 0\}$;
 4. $\{x \in \mathbf{R}^2 : x_2 = 0\}$;
 5. $\{x \in \mathbf{R}^2 : x_1 + x_2 = 0\}$;
 6. $\{x \in \mathbf{R}^2 : x_1 - x_2 = 0\}$;
 7. $\{x \in \mathbf{R}^2 : |x_1 - x_2| \leq x_1^4 + x_2^4\}$;
- Let $f : \mathbf{R}^n \rightarrow \mathbf{R}^m$ be a continuous mapping. Mark those of the following statements which always are true:

1. If U is an open set in \mathbf{R}^m , then so is the set $f^{-1}(U) = \{x : f(x) \in U\}$;
2. If U is an open set in \mathbf{R}^n , then so is the set $f(U) = \{f(x) : x \in U\}$;
3. If F is a closed set in \mathbf{R}^m , then so is the set $f^{-1}(F) = \{x : f(x) \in F\}$;
4. If F is a closed set in \mathbf{R}^n , then so is the set $f(F) = \{f(x) : x \in F\}$.

A.5.2 Elementary continuity-preserving operations

All “elementary” operations with mappings preserve continuity. Specifically,

Theorem A.5.1 *Let X be a subset in \mathbf{R}^n .*

(i) [stability of continuity w.r.t. linear operations] *If $f_1(x), f_2(x)$ are continuous functions on X taking values in \mathbf{R}^m and $\lambda_1(x), \lambda_2(x)$ are continuous real-valued functions on X , then the function*

$$f(x) = \lambda_1(x)f_1(x) + \lambda_2(x)f_2(x) : X \rightarrow \mathbf{R}^m$$

is continuous on X ;

(ii) [stability of continuity w.r.t. superposition] *Let*

- $X \subset \mathbf{R}^n, Y \subset \mathbf{R}^m$;
- $f : X \rightarrow \mathbf{R}^m$ be a continuous mapping such that $f(x) \in Y$ for every $x \in X$;
- $g : Y \rightarrow \mathbf{R}^k$ be a continuous mapping.

Then the composite mapping

$$h(x) = g(f(x)) : X \rightarrow \mathbf{R}^k$$

is continuous on X .

A.5.3 Basic properties of continuous functions on \mathbf{R}^n

The basic properties of continuous functions on \mathbf{R}^n can be summarized as follows:

Theorem A.5.2 *Let X be a nonempty closed and bounded subset of \mathbf{R}^n .*

(i) *If a mapping $f : X \rightarrow \mathbf{R}^m$ is continuous on X , it is bounded on X : there exists $C < \infty$ such that $|f(x)| \leq C$ for all $x \in X$.*

Proof. Assume, on the contrary to what should be proved, that f is unbounded, so that for every i there exists a point $x^i \in X$ such that $|f(x^i)| > i$. By Theorem A.4.2, we can extract from the sequence $\{x^i\}$ a subsequence $\{x^{i_j}\}_{j=1}^\infty$ which converges to a point $\bar{x} \in X$. The real-valued function $g(x) = |f(x)|$ is continuous (as the superposition of two continuous mappings, see Theorem A.5.1.(ii)) and therefore its values at the points x^{i_j} should converge, as $j \rightarrow \infty$, to its value at \bar{x} ; on the other hand, $g(x^{i_j}) \geq i_j \rightarrow \infty$ as $j \rightarrow \infty$, and we get the desired contradiction.

(ii) *If a mapping $f : X \rightarrow \mathbf{R}^m$ is continuous on X , it is uniformly continuous: for every $\epsilon > 0$ there exists $\delta > 0$ such that*

$$x, y \in X, |x - y| < \delta \Rightarrow |f(x) - f(y)| < \epsilon.$$

Proof. Assume, on the contrary to what should be proved, that there exists $\epsilon > 0$ such that for every $\delta > 0$ one can find a pair of points x, y in X such that $|x - y| < \delta$ and $|f(x) - f(y)| \geq \epsilon$. In particular, for every $i = 1, 2, \dots$ we can find two points x^i, y^i in X such that $|x^i - y^i| \leq 1/i$ and $|f(x^i) - f(y^i)| \geq \epsilon$. By Theorem A.4.2, we can extract from the sequence $\{x^i\}$ a subsequence $\{x^{i_j}\}_{j=1}^\infty$ which converges to certain point $\bar{x} \in X$. Since $|y^{i_j} - x^{i_j}| \leq 1/i_j \rightarrow 0$ as $j \rightarrow \infty$, the sequence $\{y^{i_j}\}_{j=1}^\infty$ converges to the same point \bar{x} as the sequence $\{x^{i_j}\}_{j=1}^\infty$ (why?). Since f is continuous, we have

$$\lim_{j \rightarrow \infty} f(y^{i_j}) = f(\bar{x}) = \lim_{j \rightarrow \infty} f(x^{i_j}),$$

whence $\lim_{j \rightarrow \infty} (f(x^{i_j}) - f(y^{i_j})) = 0$, which contradicts the fact that $|f(x^{i_j}) - f(y^{i_j})| \geq \epsilon > 0$ for all j .

(iii) Let f be a real-valued continuous function on X . The f attains its minimum on X :

$$\operatorname{Argmin}_X f \equiv \{x \in X : f(x) = \inf_{y \in X} f(y)\} \neq \emptyset,$$

same as f attains its maximum at certain points of X :

$$\operatorname{Argmax}_X f \equiv \{x \in X : f(x) = \sup_{y \in X} f(y)\} \neq \emptyset.$$

Proof: Let us prove that f attains its maximum on X (the proof for minimum is completely similar). Since f is bounded on X by (i), the quantity

$$f^* = \sup_{x \in X} f(x)$$

is finite; of course, we can find a sequence $\{x^i\}$ of points from X such that $f^* = \lim_{i \rightarrow \infty} f(x^i)$. By Theorem A.4.2, we can extract from the sequence $\{x^i\}$ a subsequence $\{x^{i_j}\}_{j=1}^\infty$ which converges to certain point $\bar{x} \in X$. Since f is continuous on X , we have

$$f(\bar{x}) = \lim_{j \rightarrow \infty} f(x^{i_j}) = \lim_{i \rightarrow \infty} f(x^i) = f^*,$$

so that the maximum of f on X indeed is achieved (e.g., at the point \bar{x}).

Exercise A.6 Prove that in general no one of the three statements in Theorem A.5.2 remains valid when X is closed, but not bounded, same as when X is bounded, but not closed.

A.6 Differentiable functions on \mathbf{R}^n

A.6.1 The derivative

The reader definitely is familiar with the notion of derivative of a real-valued function $f(x)$ of real variable x :

$$f'(x) = \lim_{\Delta x \rightarrow 0} \frac{f(x + \Delta x) - f(x)}{\Delta x}$$

This definition does not work when we pass from functions of single real variable to functions of several real variables, or, which is the same, to functions with vector arguments. Indeed, in this case the shift in the argument Δx should be a vector, and we do not know what does it mean to divide by a vector...

A proper way to extend the notion of the derivative to real- and vector-valued functions of vector argument is to realize what in fact is the meaning of the derivative in the univariate case. What $f'(x)$ says to us is *how to approximate f in a neighbourhood of x by a linear function*. Specifically, if $f'(x)$ exists, then the linear function $f'(x)\Delta x$ of Δx approximates the change $f(x + \Delta x) - f(x)$ in f up to a remainder which is of highest order as compared with Δx as $\Delta x \rightarrow 0$:

$$|f(x + \Delta x) - f(x) - f'(x)\Delta x| \leq \bar{o}(|\Delta x|) \text{ as } \Delta x \rightarrow 0.$$

In the above formula, we meet with the notation $\bar{o}(|\Delta x|)$, and here is the explanation of this notation:

$\bar{o}(|\Delta x|)$ is a common name of all functions $\phi(\Delta x)$ of Δx which are well-defined in a neighbourhood of the point $\Delta x = 0$ on the axis, vanish at the point $\Delta x = 0$ and are such that

$$\frac{\phi(\Delta x)}{|\Delta x|} \rightarrow 0 \text{ as } \Delta x \rightarrow 0.$$

For example,

1. $(\Delta x)^2 = \bar{o}(|\Delta x|)$, $\Delta x \rightarrow 0$,
2. $|\Delta x|^{1.01} = \bar{o}(|\Delta x|)$, $\Delta x \rightarrow 0$,
3. $\sin^2(\Delta x) = \bar{o}(|\Delta x|)$, $\Delta x \rightarrow 0$,
4. $\Delta x \neq \bar{o}(|\Delta x|)$, $\Delta x \rightarrow 0$.

Later on we shall meet with the notation “ $\bar{o}(|\Delta x|^k)$ as $\Delta x \rightarrow 0$ ”, where k is a positive integer. The definition is completely similar to the one for the case of $k = 1$:

$\bar{o}(|\Delta x|^k)$ is a common name of all functions $\phi(\Delta x)$ of Δx which are well-defined in a neighbourhood of the point $\Delta x = 0$ on the axis, vanish at the point $\Delta x = 0$ and are such that

$$\frac{\phi(\Delta x)}{|\Delta x|^k} \rightarrow 0 \text{ as } \Delta x \rightarrow 0.$$

Note that if $f(\cdot)$ is a function defined in a neighbourhood of a point x on the axis, then there perhaps are many linear functions $a\Delta x$ of Δx which well approximate $f(x + \Delta x) - f(x)$, in the sense that the remainder in the approximation

$$f(x + \Delta x) - f(x) - a\Delta x$$

tends to 0 as $\Delta x \rightarrow 0$; among these approximations, however, there exists *at most one* which approximates $f(x + \Delta x) - f(x)$ “very well” – so that the remainder is $\bar{o}(|\Delta x|)$, and not merely tends to 0 as $\Delta x \rightarrow 0$. Indeed, if

$$f(x + \Delta x) - f(x) - a\Delta x = \bar{o}(|\Delta x|),$$

then, dividing both sides by Δx , we get

$$\frac{f(x + \Delta x) - f(x)}{\Delta x} - a = \frac{\bar{o}(|\Delta x|)}{\Delta x};$$

by definition of $\bar{o}(\cdot)$, the right hand side in this equality tends to 0 as $\Delta x \rightarrow 0$, whence

$$a = \lim_{\Delta x \rightarrow 0} \frac{f(x + \Delta x) - f(x)}{\Delta x} = f'(x).$$

Thus, if a linear function $a\Delta x$ of Δx approximates the change $f(x + \Delta x) - f(x)$ in f up to the remainder which is $\bar{o}(|\Delta x|)$ as $\Delta x \rightarrow 0$, then a is the derivative of f at x . You can easily verify that the inverse statement also is true: if the derivative of f at x exists, then the linear function $f'(x)\Delta x$ of Δx approximates the change $f(x + \Delta x) - f(x)$ in f up to the remainder which is $\bar{o}(|\Delta x|)$ as $\Delta x \rightarrow 0$.

The advantage of the “ $\bar{o}(|\Delta x|)$ ”-definition of derivative is that it can be naturally extended onto vector-valued functions of vector arguments (you should just replace “axis” with \mathbf{R}^n in the definition of \bar{o}) and enlightens the essence of the notion of derivative: when it exists, this is exactly *the linear function of Δx which approximates the change $f(x + \Delta x) - f(x)$ in f up to a remainder which is $\bar{o}(|\Delta x|)$* . The precise definition is as follows:

Definition A.6.1 [Frechet differentiability] Let f be a function which is well-defined in a neighbourhood of a point $x \in \mathbf{R}^n$ and takes values in \mathbf{R}^m . We say that f is differentiable at x , if there exists a linear function $Df(x)[\Delta x]$ of $\Delta x \in \mathbf{R}^n$ taking values in \mathbf{R}^m which approximates the change $f(x + \Delta x) - f(x)$ in f up to a remainder which is $\bar{o}(|\Delta x|)$:

$$|f(x + \Delta x) - f(x) - Df(x)[\Delta x]| \leq \bar{o}(|\Delta x|). \quad (\text{A.6.1})$$

Equivalently: a function f which is well-defined in a neighbourhood of a point $x \in \mathbf{R}^n$ and takes values in \mathbf{R}^m is called differentiable at x , if there exists a linear function $Df(x)[\Delta x]$ of $\Delta x \in \mathbf{R}^n$ taking values in \mathbf{R}^m such that for every $\epsilon > 0$ there exists $\delta > 0$ satisfying the relation

$$|\Delta x| \leq \delta \Rightarrow |f(x + \Delta x) - f(x) - Df(x)[\Delta x]| \leq \epsilon |\Delta x|.$$

A.6.2 Derivative and directional derivatives

We have defined what does it mean that a function $f : \mathbf{R}^n \rightarrow \mathbf{R}^m$ is differentiable at a point x , but did not say yet what is the derivative. The reader could guess that the derivative is exactly the “linear function $Df(x)[\Delta x]$ of $\Delta x \in \mathbf{R}^n$ taking values in \mathbf{R}^m which approximates the change $f(x + \Delta x) - f(x)$ in f up to a remainder which is $\leq \bar{o}(|\Delta x|)$ ” participating in the definition of differentiability. The guess is correct, but we cannot merely call the entity participating in the definition the derivative – why do we know that this entity is unique? Perhaps there are many different linear functions of Δx approximating the change in f up to a remainder which is $\bar{o}(|\Delta x|)$. In fact there is no more than a single linear function with this property due to the following observation:

Let f be differentiable at x , and $Df(x)[\Delta x]$ be a linear function participating in the definition of differentiability. Then

$$\forall \Delta x \in \mathbf{R}^n : Df(x)[\Delta x] = \lim_{t \rightarrow +0} \frac{f(x + t\Delta x) - f(x)}{t}. \quad (\text{A.6.2})$$

In particular, the derivative $Df(x)[\cdot]$ is uniquely defined by f and x .

Proof. We have

$$\begin{aligned} & |f(x + t\Delta x) - f(x) - Df(x)[t\Delta x]| \leq \bar{o}(|t\Delta x|) \\ & \quad \downarrow \\ & \left| \frac{f(x + t\Delta x) - f(x)}{t} - \frac{Df(x)[t\Delta x]}{t} \right| \leq \frac{\bar{o}(|t\Delta x|)}{t} \\ & \quad \Updownarrow \\ & \left| \frac{f(x + t\Delta x) - f(x)}{t} - Df(x)[\Delta x] \right| \leq \frac{\bar{o}(|t\Delta x|)}{t} \quad [\text{since } Df(x)[\cdot] \text{ is linear}] \\ & \quad \Downarrow \\ & Df(x)[\Delta x] = \lim_{t \rightarrow +0} \frac{f(x + t\Delta x) - f(x)}{t} \quad \left[\begin{array}{l} \text{passing to limit as } t \rightarrow +0; \\ \text{note that } \frac{\bar{o}(|t\Delta x|)}{t} \rightarrow 0, t \rightarrow +0 \end{array} \right] \end{aligned}$$

Pay attention to important remarks as follows:

1. The right hand side limit in (A.6.2) is an important entity called the *directional derivative of f taken at x along (a direction) Δx* ; note that this quantity is defined in the “purely univariate” fashion – by dividing the change in f by the magnitude of a shift in a direction Δx and passing to limit as the magnitude of the shift approaches 0. Relation (A.6.2) says that the derivative, if exists, is, at every Δx , nothing that the directional derivative of f taken at x along Δx . Note, however, that differentiability is much more than the existence of directional derivatives along all directions Δx ; differentiability requires also *the directional derivatives to be “well-organized” – to depend linearly on the direction Δx* . It is easily seen that just existence of directional derivatives does not imply their “good organization”: for example, the Euclidean norm

$$f(x) = |x|$$

at $x = 0$ possesses directional derivatives along all directions:

$$\lim_{t \rightarrow +0} \frac{f(0 + t\Delta x) - f(0)}{t} = |\Delta x|;$$

these derivatives, however, depend *non-linearly* on Δx , so that the Euclidean norm is *not* differentiable at the origin (although is differentiable everywhere outside the origin, but this is another story).

2. It should be stressed that the derivative, if exists, is what it is: *a linear function of $\Delta x \in \mathbf{R}^n$ taking values in \mathbf{R}^m* . As we shall see in a while, we can represent this function by something “tractable”, like a vector or a matrix, and can understand how to compute such a representation; however, an intelligent reader should bear in mind that a representation is not exactly the same as *the* represented entity. Sometimes the difference between derivatives and the entities which represent them is reflected in the terminology: what we call the *derivative*, is also called the *differential*, while the word “derivative” is reserved for the vector/matrix representing the differential.

A.6.3 Representations of the derivative

indexderivatives!representation ofBy definition, the derivative of a mapping $f : \mathbf{R}^n \rightarrow \mathbf{R}^m$ at a point x is a linear function $Df(x)[\Delta x]$ taking values in \mathbf{R}^m . How could we represent such a function?

Case of $m = 1$ – the gradient. Let us start with real-valued functions (i.e., with the case of $m = 1$); in this case the derivative is a *linear* real-valued function on \mathbf{R}^n . As we remember, the standard Euclidean structure on \mathbf{R}^n allows to represent every linear function on \mathbf{R}^n as the inner product of the argument with certain fixed vector. In particular, the derivative $Df(x)[\Delta x]$ of a scalar function can be represented as

$$Df(x)[\Delta x] = [\text{vector}]^T \Delta x;$$

what is denoted “vector” in this relation, is called the *gradient* of f at x and is denoted by $\nabla f(x)$:

$$Df(x)[\Delta x] = (\nabla f(x))^T \Delta x. \quad (\text{A.6.3})$$

How to compute the gradient? The answer is given by (A.6.2). Indeed, let us look what (A.6.3) and (A.6.2) say when Δx is the i -th standard basic orth. According to (A.6.3), $Df(x)[e_i]$ is the i -th coordinate of the vector $\nabla f(x)$; according to (A.6.2),

$$\left. \begin{aligned} Df(x)[e_i] &= \lim_{t \rightarrow +0} \frac{f(x+te_i) - f(x)}{t}, \\ Df(x)[e_i] = -Df(x)[-e_i] &= -\lim_{t \rightarrow +0} \frac{f(x-te_i) - f(x)}{t} = \lim_{t \rightarrow -0} \frac{f(x+te_i) - f(x)}{t} \end{aligned} \right\} \Rightarrow Df(x)[e_i] = \frac{\partial f(x)}{\partial x_i}.$$

Thus,

If a real-valued function f is differentiable at x , then the first order partial derivatives of f at x exist, and the gradient of f at x is just the vector with the coordinates which are the first order partial derivatives of f taken at x :

$$\nabla f(x) = \begin{bmatrix} \frac{\partial f(x)}{\partial x_1} \\ \vdots \\ \frac{\partial f(x)}{\partial x_n} \end{bmatrix}.$$

The derivative of f , taken at x , is the linear function of Δx given by

$$Df(x)[\Delta x] = (\nabla f(x))^T \Delta x = \sum_{i=1}^n \frac{\partial f(x)}{\partial x_i} (\Delta x)_i.$$

General case – the Jacobian. Now let $f : \mathbf{R}^n \rightarrow \mathbf{R}^m$ with $m \geq 1$. In this case, $Df(x)[\Delta x]$, regarded as a function of Δx , is a linear mapping from \mathbf{R}^n to \mathbf{R}^m ; as we remember, the standard way to represent a linear mapping from \mathbf{R}^n to \mathbf{R}^m is to represent it as the multiplication by $m \times n$ matrix:

$$Df(x)[\Delta x] = [m \times n \text{ matrix}] \cdot \Delta x. \quad (\text{A.6.4})$$

What is denoted by “matrix” in (A.6.4), is called the *Jacobian* of f at x and is denoted by $f'(x)$. How to compute the entries of the Jacobian? Here again the answer is readily given by (A.6.2). Indeed, on one hand, we have

$$Df(x)[\Delta x] = f'(x)\Delta x, \quad (\text{A.6.5})$$

whence

$$[Df(x)[e_j]]_i = (f'(x))_{ij}, \quad i = 1, \dots, m, j = 1, \dots, n.$$

On the other hand, denoting

$$f(x) = \begin{bmatrix} f_1(x) \\ \vdots \\ f_m(x) \end{bmatrix},$$

the same computation as in the case of gradient demonstrates that

$$[Df(x)[e_j]]_i = \frac{\partial f_i(x)}{\partial x_j}$$

and we arrive at the following conclusion:

If a vector-valued function $f(x) = (f_1(x), \dots, f_m(x))$ is differentiable at x , then the first order partial derivatives of all f_i at x exist, and the Jacobian of f at x is just the $m \times n$ matrix with the entries $[\frac{\partial f_i(x)}{\partial x_j}]_{i,j}$ (so that the rows in the Jacobian are $[\nabla f_1(x)]^T, \dots, [\nabla f_m(x)]^T$). The derivative of f , taken at x , is the linear vector-valued function of Δx given by

$$Df(x)[\Delta x] = f'(x)\Delta x = \begin{bmatrix} [\nabla f_1(x)]^T \Delta x \\ \vdots \\ [\nabla f_m(x)]^T \Delta x \end{bmatrix}.$$

Remark A.6.1 Note that for a real-valued function f we have defined both the gradient $\nabla f(x)$ and the Jacobian $f'(x)$. These two entities are “nearly the same”, but not exactly the same: the Jacobian is a vector-row, and the gradient is a vector-column linked by the relation

$$f'(x) = (\nabla f(x))^T.$$

Of course, both these representations of the derivative of f yield the same linear approximation of the change in f :

$$Df(x)[\Delta x] = (\nabla f(x))^T \Delta x = f'(x)\Delta x.$$

A.6.4 Existence of the derivative

We have seen that the existence of the derivative of f at a point implies the existence of the first order partial derivatives of the (components f_1, \dots, f_m of) f . The inverse statement is not exactly true – the existence of all first order partial derivatives $\frac{\partial f_i(x)}{\partial x_j}$ not necessarily implies the existence of the derivative; we need a bit more:

Theorem A.6.1 [Sufficient condition for differentiability] *Assume that*

1. *The mapping $f = (f_1, \dots, f_m) : \mathbf{R}^n \rightarrow \mathbf{R}^m$ is well-defined in a neighbourhood U of a point $x_0 \in \mathbf{R}^n$,*
2. *The first order partial derivatives of the components f_i of f exist everywhere in U , and*
3. *The first order partial derivatives of the components f_i of f are continuous at the point x_0 .*

Then f is differentiable at the point x_0 .

A.6.5 Calculus of derivatives

The calculus of derivatives is given by the following result:

Theorem A.6.2 (i) [Differentiability and linear operations] *Let $f_1(x), f_2(x)$ be mappings defined in a neighbourhood of $x_0 \in \mathbf{R}^n$ and taking values in \mathbf{R}^m , and $\lambda_1(x), \lambda_2(x)$ be real-valued*

functions defined in a neighbourhood of x_0 . Assume that $f_1, f_2, \lambda_1, \lambda_2$ are differentiable at x_0 . Then so is the function $f(x) = \lambda_1(x)f_1(x) + \lambda_2(x)f_2(x)$, with the derivative at x_0 given by

$$\begin{aligned} Df(x_0)[\Delta x] &= [D\lambda_1(x_0)[\Delta x]]f_1(x_0) + \lambda_1(x_0)Df_1(x_0)[\Delta x] \\ &\quad + [D\lambda_2(x_0)[\Delta x]]f_2(x_0) + \lambda_2(x_0)Df_2(x_0)[\Delta x] \\ &\quad \Downarrow \\ f'(x_0) &= f_1(x_0)[\nabla\lambda_1(x_0)]^T + \lambda_1(x_0)f'_1(x_0) \\ &\quad + f_2(x_0)[\nabla\lambda_2(x_0)]^T + \lambda_2(x_0)f'_2(x_0). \end{aligned}$$

(ii) [chain rule] Let a mapping $f : \mathbf{R}^n \rightarrow \mathbf{R}^m$ be differentiable at x_0 , and a mapping $g : \mathbf{R}^m \rightarrow \mathbf{R}^n$ be differentiable at $y_0 = f(x_0)$. Then the superposition $h(x) = g(f(x))$ is differentiable at x_0 , with the derivative at x_0 given by

$$\begin{aligned} Dh(x_0)[\Delta x] &= Dg(y_0)[Df(x_0)[\Delta x]] \\ &\quad \Downarrow \\ h'(x_0) &= g'(y_0)f'(x_0) \end{aligned}$$

If the outer function g is real-valued, then the latter formula implies that

$$\nabla h(x_0) = [f'(x_0)]^T \nabla g(y_0)$$

(recall that for a real-valued function ϕ , $\phi' = (\nabla\phi)^T$).

A.6.6 Computing the derivative

Representations of the derivative via first order partial derivatives normally allow to compute it by the standard Calculus rules, in a completely mechanical fashion, not thinking at all of what we are computing. The examples to follow (especially the third of them) demonstrate that it often makes sense to bear in mind what is the derivative; this sometimes yield the result much faster than blind implementing Calculus rules.

Example 1: The gradient of an affine function. An affine function

$$f(x) = a + \sum_{i=1}^n g_i x_i \equiv a + g^T x : \mathbf{R}^n \rightarrow \mathbf{R}$$

is differentiable at every point (Theorem A.6.1) and its gradient, of course, equals g :

$$\begin{aligned} (\nabla f(x))^T \Delta x &= \lim_{t \rightarrow +0} t^{-1} [f(x + t\Delta x) - f(x)] && [(A.6.2)] \\ &= \lim_{t \rightarrow +0} t^{-1} [tg^T \Delta x] && [\text{arithmetics}] \end{aligned}$$

and we arrive at

$$\boxed{\nabla(a + g^T x) = g}$$

Example 2: The gradient of a quadratic form. For the time being, let us define a homogeneous quadratic form on \mathbf{R}^n as a function

$$f(x) = \sum_{i,j} A_{ij} x_i x_j = x^T A x,$$

where A is an $n \times n$ matrix. Note that the matrices A and A^T define the same quadratic form, and therefore the symmetric matrix $B = \frac{1}{2}(A + A^T)$ also produces the same quadratic form as A and A^T . It follows that we always may assume (and do assume from now on) that the matrix A producing the quadratic form in question is symmetric.

A quadratic form is a simple polynomial and as such is differentiable at every point (Theorem A.6.1). What is the gradient of f at a point x ? Here is the computation:

$$\begin{aligned}
(\nabla f(x))^T \Delta x &= Df(x)[\Delta x] \\
&= \lim_{t \rightarrow +0} [(x + t\Delta x)^T A(x + t\Delta x) - x^T Ax] \quad [(\text{A.6.2})] \\
&= \lim_{t \rightarrow +0} [x^T Ax + t(\Delta x)^T Ax + tx^T A\Delta x + t^2(\Delta x)^T A\Delta x - x^T Ax] \quad [\text{opening parentheses}] \\
&= \lim_{t \rightarrow +0} t^{-1} [2t(Ax)^T \Delta x + t^2(\Delta x)^T A\Delta x] \quad [\text{arithmetics + symmetry of } A] \\
&= 2(Ax)^T \Delta x
\end{aligned}$$

We conclude that

$$\boxed{\nabla(x^T Ax) = 2Ax}$$

(recall that $A = A^T$).

Example 3: The derivative of the log-det barrier. Let us compute the derivative of the *log-det barrier* (playing an extremely important role in modern optimization)

$$F(X) = \ln \text{Det}(X);$$

here X is an $n \times n$ matrix (or, if you prefer, n^2 -dimensional vector). Note that $F(X)$ is well-defined and differentiable in a neighbourhood of every point \bar{X} with positive determinant (indeed, $\text{Det}(X)$ is a polynomial of the entries of X and thus – is everywhere continuous and differentiable with continuous partial derivatives, while the function $\ln(t)$ is continuous and differentiable on the positive ray; by Theorems A.5.1.(ii), A.6.2.(ii), F is differentiable at every X such that $\text{Det}(X) > 0$). The reader is kindly asked to try to find the derivative of F by the standard techniques; if the result will not be obtained in, say, 30 minutes, please look at the 8-line computation to follow (in this computation, $\text{Det}(\bar{X}) > 0$, and $G(X) = \text{Det}(X)$):

$$\begin{aligned}
&DF(\bar{X})[\Delta X] \\
&= D\ln(G(\bar{X}))[DG(\bar{X})[\Delta X]] \quad [\text{chain rule}] \\
&= G^{-1}(\bar{X})DG(\bar{X})[\Delta X] \quad [\ln'(t) = t^{-1}] \\
&= \text{Det}^{-1}(\bar{X}) \lim_{t \rightarrow +0} t^{-1} [\text{Det}(\bar{X} + t\Delta X) - \text{Det}(\bar{X})] \quad [\text{definition of } G \text{ and (A.6.2)}] \\
&= \text{Det}^{-1}(\bar{X}) \lim_{t \rightarrow +0} t^{-1} [\text{Det}(\bar{X}(I + t\bar{X}^{-1}\Delta X)) - \text{Det}(\bar{X})] \\
&= \text{Det}^{-1}(\bar{X}) \lim_{t \rightarrow +0} t^{-1} [\text{Det}(\bar{X})(\text{Det}(I + t\bar{X}^{-1}\Delta X) - 1)] \quad [\text{Det}(AB) = \text{Det}(A)\text{Det}(B)] \\
&= \lim_{t \rightarrow +0} t^{-1} [\text{Det}(I + t\bar{X}^{-1}\Delta X) - 1] \\
&= \text{Tr}(\bar{X}^{-1}\Delta X) = \sum_{i,j} [\bar{X}^{-1}]_{ji} (\Delta X)_{ij}
\end{aligned}$$

where the concluding equality

$$\lim_{t \rightarrow +0} t^{-1} [\text{Det}(I + tA) - 1] = \text{Tr}(A) \equiv \sum_i A_{ii} \quad (\text{A.6.6})$$

is immediately given by recalling what is the determinant of $I + tA$: this is a polynomial of t which is the sum of products, taken along all diagonals of a $n \times n$ matrix and assigned certain signs, of the entries of $I + tA$. At every one of these diagonals, except for the main one, there are at least two cells with the entries proportional to t , so that the corresponding products do not contribute to the constant and the linear in t terms in $\text{Det}(I + tA)$ and thus do not affect the limit in (A.6.6). The only product which does contribute to the linear and the constant terms in $\text{Det}(I + tA)$ is the product $(1+tA_{11})(1+tA_{22})\dots(1+tA_{nn})$ coming from the main diagonal; it is clear that in this product the constant term is 1, and the linear in t term is $t(A_{11} + \dots + A_{nn})$, and (A.6.6) follows.

A.6.7 Higher order derivatives

Let $f : \mathbf{R}^n \rightarrow \mathbf{R}^m$ be a mapping which is well-defined and differentiable at every point x from an open set U . The Jacobian of this mapping $J(x)$ is a mapping from \mathbf{R}^n to the space $\mathbf{R}^{m \times n}$ matrices, i.e., is a mapping taking values in certain \mathbf{R}^M ($M = mn$). The derivative of this mapping, if it exists, is called the *second derivative* of f ; it again is a mapping from \mathbf{R}^n to certain \mathbf{R}^M and as such can be differentiable, and so on, so that we can speak about the second, the third, ... derivatives of a vector-valued function of vector argument. A *sufficient* condition for the existence of k derivatives of f in U is that f is C^k in U , i.e., that all partial derivatives of f of orders $\leq k$ exist and are continuous everywhere in U (cf. Theorem A.6.1).

We have explained what does it mean that f has k derivatives in U ; note, however, that according to the definition, highest order derivatives at a point x are just long vectors; say, the second order derivative of a scalar function f of 2 variables is the Jacobian of the mapping $x \mapsto f'(x) : \mathbf{R}^2 \rightarrow \mathbf{R}^2$, i.e., a mapping from \mathbf{R}^2 to $\mathbf{R}^{2 \times 2} = \mathbf{R}^4$; the third order derivative of f is therefore the Jacobian of a mapping from \mathbf{R}^2 to \mathbf{R}^4 , i.e., a mapping from \mathbf{R}^2 to $\mathbf{R}^{4 \times 2} = \mathbf{R}^8$, and so on. The question which should be addressed now is: *What is a natural and transparent way to represent the highest order derivatives?*

The answer is as follows:

(*) Let $f : \mathbf{R}^n \rightarrow \mathbf{R}^m$ be C^k on an open set $U \subset \mathbf{R}^n$. The derivative of order $\ell \leq k$ of f , taken at a point $x \in U$, can be naturally identified with a function

$$D^\ell f(x)[\Delta x^1, \Delta x^2, \dots, \Delta x^\ell]$$

of ℓ vector arguments $\Delta x^i \in \mathbf{R}^n$, $i = 1, \dots, \ell$, and taking values in \mathbf{R}^m . This function is linear in every one of the arguments Δx^i , the other arguments being fixed, and is symmetric with respect to permutation of arguments $\Delta x^1, \dots, \Delta x^\ell$.

In terms of f , the quantity $D^\ell f(x)[\Delta x^1, \Delta x^2, \dots, \Delta x^\ell]$ (full name: “the ℓ -th derivative (or differential) of f taken at a point x along the directions $\Delta x^1, \dots, \Delta x^\ell$ ”) is given by

$$D^\ell f(x)[\Delta x^1, \Delta x^2, \dots, \Delta x^\ell] = \frac{\partial^\ell}{\partial t_\ell \partial t_{\ell-1} \dots \partial t_1} \Big|_{t_1=\dots=t_\ell=0} f(x+t_1 \Delta x^1 + t_2 \Delta x^2 + \dots + t_\ell \Delta x^\ell). \quad (\text{A.6.7})$$

The explanation to our claims is as follows. Let $f : \mathbf{R}^n \rightarrow \mathbf{R}^m$ be C^k on an open set $U \subset \mathbf{R}^n$.

1. When $\ell = 1$, (*) says to us that the first order derivative of f , taken at x , is a linear function $Df(x)[\Delta x^1]$ of $\Delta x^1 \in \mathbf{R}^n$, taking values in \mathbf{R}^m , and that the value of this function at every

Δx^1 is given by the relation

$$Df(x)[\Delta x^1] = \frac{\partial}{\partial t_1} \Big|_{t_1=0} f(x + t_1 \Delta x^1) \quad (\text{A.6.8})$$

(cf. (A.6.2)), which is in complete accordance with what we already know about the derivative.

2. To understand what is the second derivative, let us take the first derivative $Df(x)[\Delta x^1]$, *let us temporarily fix somehow the argument Δx^1* and treat the derivative as a function of x . As a function of x , Δx^1 being fixed, the quantity $Df(x)[\Delta x^1]$ is again a mapping which maps U into \mathbf{R}^m and is differentiable by Theorem A.6.1 (provided, of course, that $k \geq 2$). The derivative of this mapping is certain linear function of $\Delta x \equiv \Delta x^2 \in \mathbf{R}^n$, depending on x as on a parameter; and of course it depends on Δx^1 as on a parameter as well. Thus, the derivative of $Df(x)[\Delta x^1]$ in x is certain function

$$D^2 f(x)[\Delta x^1, \Delta x^2]$$

of $x \in U$ and $\Delta x^1, \Delta x^2 \in \mathbf{R}^n$ and taking values in \mathbf{R}^m . What we know about this function is that it is linear in Δx^2 . In fact, it is also linear in Δx^1 , since it is the derivative in x of certain function (namely, of $Df(x)[\Delta x^1]$) *linearly depending on the parameter Δx^1* , so that the derivative of the function *in* x is linear in the parameter Δx^1 as well (differentiation is a linear operation with respect to a function we are differentiating: summing up functions and multiplying them by real constants, we sum up, respectively, multiply by the same constants, the derivatives). Thus, $D^2 f(x)[\Delta x^1, \Delta x^2]$ is linear in Δx^1 when x and Δx^2 are fixed, and is linear in Δx^2 when x and Δx^1 are fixed. Moreover, we have

$$\begin{aligned} D^2 f(x)[\Delta x^1, \Delta x^2] &= \frac{\partial}{\partial t_2} \Big|_{t_2=0} Df(x + t_2 \Delta x^2)[\Delta x^1] && [\text{cf. (A.6.8)}] \\ &= \frac{\partial}{\partial t_2} \Big|_{t_2=0} \frac{\partial}{\partial t_1} \Big|_{t_1=0} f(x + t_2 \Delta x^2 + t_1 \Delta x^1) && [\text{by (A.6.8)}] \\ &= \frac{\partial^2}{\partial t_2 \partial t_1} \Big|_{t_1=t_2=0} f(x + t_1 \Delta x^1 + t_2 \Delta x^2) \end{aligned} \quad (\text{A.6.9})$$

as claimed in (A.6.7) for $\ell = 2$. The only piece of information about the second derivative which is contained in (*) and is not justified yet is that $D^2 f(x)[\Delta x^1, \Delta x^2]$ is symmetric in $\Delta x^1, \Delta x^2$; but this fact is readily given by the representation (A.6.7), since, as they prove in Calculus, if a function ϕ possesses *continuous* partial derivatives of orders $\leq \ell$ in a neighbourhood of a point, then these derivatives in this neighbourhood are independent of the order in which they are taken; it follows that

$$\begin{aligned} D^2 f(x)[\Delta x^1, \Delta x^2] &= \frac{\partial^2}{\partial t_2 \partial t_1} \Big|_{t_1=t_2=0} \underbrace{f(x + t_1 \Delta x^1 + t_2 \Delta x^2)}_{\phi(t_1, t_2)} && [(\text{A.6.9})] \\ &= \frac{\partial^2}{\partial t_1 \partial t_2} \Big|_{t_1=t_2=0} \phi(t_1, t_2) \\ &= \frac{\partial^2}{\partial t_1 \partial t_2} \Big|_{t_1=t_2=0} f(x + t_2 \Delta x^2 + t_1 \Delta x^1) \\ &= D^2 f(x)[\Delta x^2, \Delta x^1] && [\text{the same (A.6.9)}] \end{aligned}$$

3. Now it is clear how to proceed: to define $D^3 f(x)[\Delta x^1, \Delta x^2, \Delta x^3]$, we fix in the second order derivative $D^2 f(x)[\Delta x^1, \Delta x^2]$ the arguments $\Delta x^1, \Delta x^2$ and treat it as a function of

x only, thus arriving at a mapping which maps U into \mathbf{R}^m and depends on $\Delta x^1, \Delta x^2$ as on parameters (linearly in every one of them). Differentiating the resulting mapping in x , we arrive at a function $D^3 f(x)[\Delta x^1, \Delta x^2, \Delta x^3]$ which by construction is linear in every one of the arguments $\Delta x^1, \Delta x^2, \Delta x^3$ and satisfies (A.6.7); the latter relation, due to the Calculus result on the symmetry of partial derivatives, implies that $D^3 f(x)[\Delta x^1, \Delta x^2, \Delta x^3]$ is symmetric in $\Delta x^1, \Delta x^2, \Delta x^3$. After we have at our disposal the third derivative $D^3 f$, we can build from it in the already explained fashion the fourth derivative, and so on, until k -th derivative is defined.

Remark A.6.2 Since $D^\ell f(x)[\Delta x^1, \dots, \Delta x^\ell]$ is linear in every one of Δx^i , we can expand the derivative in a multiple sum:

$$\begin{aligned} \Delta x^i &= \sum_{j=1}^n \Delta x_j^i e_j \\ \downarrow \\ D^\ell f(x)[\Delta x^1, \dots, \Delta x^\ell] &= D^\ell f(x) \left[\sum_{j_1=1}^n \Delta x_{j_1}^1 e_{j_1}, \dots, \sum_{j_\ell=1}^n \Delta x_{j_\ell}^\ell e_{j_\ell} \right] \\ &= \sum_{1 \leq j_1, \dots, j_\ell \leq n} D^\ell f(x)[e_{j_1}, \dots, e_{j_\ell}] \Delta x_{j_1}^1 \dots \Delta x_{j_\ell}^\ell \end{aligned} \quad (\text{A.6.10})$$

What is the origin of the coefficients $D^\ell f(x)[e_{j_1}, \dots, e_{j_\ell}]$? According to (A.6.7), one has

$$\begin{aligned} D^\ell f(x)[e_{j_1}, \dots, e_{j_\ell}] &= \frac{\partial^\ell}{\partial t_\ell \partial t_{\ell-1} \dots \partial t_1} \Big|_{t_1=\dots=t_\ell=0} f(x + t_1 e_{j_1} + t_2 e_{j_2} + \dots + t_\ell e_{j_\ell}) \\ &= \frac{\partial^\ell}{\partial x_{j_\ell} \partial x_{j_{\ell-1}} \dots \partial x_{j_1}} f(x). \end{aligned}$$

so that the coefficients in (A.6.10) are nothing but the partial derivatives, of order ℓ , of f .

Remark A.6.3 An important particular case of relation (A.6.7) is the one when $\Delta x^1 = \Delta x^2 = \dots = \Delta x^\ell$; let us call the common value of these ℓ vectors d . According to (A.6.7), we have

$$D^\ell f(x)[d, d, \dots, d] = \frac{\partial^\ell}{\partial t_\ell \partial t_{\ell-1} \dots \partial t_1} \Big|_{t_1=\dots=t_\ell=0} f(x + t_1 d + t_2 d + \dots + t_\ell d).$$

This relation can be interpreted as follows: consider the function

$$\phi(t) = f(x + td)$$

of a real variable t . Then (check it!)

$$\phi^{(\ell)}(0) = \frac{\partial^\ell}{\partial t_\ell \partial t_{\ell-1} \dots \partial t_1} \Big|_{t_1=\dots=t_\ell=0} f(x + t_1 d + t_2 d + \dots + t_\ell d) = D^\ell f(x)[d, \dots, d].$$

In other words, $D^\ell f(x)[d, \dots, d]$ is what is called ℓ -th directional derivative of f taken at x along the direction d ; to define this quantity, we pass from function f of several variables to the univariate function $\phi(t) = f(x + td)$ – restrict f onto the line passing through x and directed by d – and then take the “usual” derivative of order ℓ of the resulting function of single real variable t at the point $t = 0$ (which corresponds to the point x of our line).

Representation of higher order derivatives. k -th order derivative $D^k f(x)[\cdot, \dots, \cdot]$ of a C^k function $f : \mathbf{R}^n \rightarrow \mathbf{R}^m$ is what it is – it is a symmetric k -linear mapping on \mathbf{R}^n taking values in \mathbf{R}^m and depending on x as on a parameter. Choosing somehow coordinates in \mathbf{R}^n , we can represent such a mapping in the form

$$D^k f(x)[\Delta x_1, \dots, \Delta x_k] = \sum_{1 \leq i_1, \dots, i_k \leq n} \frac{\partial^k f(x)}{\partial x_{i_k} \partial x_{i_{k-1}} \dots \partial x_{i_1}} (\Delta x_1)_{i_1} \dots (\Delta x_k)_{i_k}.$$

We may say that the derivative can be represented by k -index collection of m -dimensional vectors $\frac{\partial^k f(x)}{\partial x_{i_k} \partial x_{i_{k-1}} \dots \partial x_{i_1}}$. This collection, however, is a difficult-to-handle entity, so that such a representation does not help. There is, however, a case when the collection becomes an entity we know to handle; this is the case of the second-order derivative of a scalar function ($k = 2, m = 1$). In this case, the collection in question is just a symmetric matrix $H(x) = \left[\frac{\partial^2 f(x)}{\partial x_i \partial x_j} \right]_{1 \leq i, j \leq n}$. This matrix is called the *Hessian* of f at x . Note that

$$D^2 f(x)[\Delta x_1, \Delta x_2] = \Delta x_1^T H(x) \Delta x_2.$$

A.6.8 Calculus of C^k mappings

The calculus of C^k mappings can be summarized as follows:

Theorem A.6.3 (i) Let U be an open set in \mathbf{R}^n , $f_1(\cdot), f_2(\cdot) : \mathbf{R}^n \rightarrow \mathbf{R}^m$ be C^k in U , and let real-valued functions $\lambda_1(\cdot), \lambda_2(\cdot)$ be C^k in U . Then the function

$$f(x) = \lambda_1(x)f_1(x) + \lambda_2(x)f_2(x)$$

is C^k in U .

(ii) Let U be an open set in \mathbf{R}^n , V be an open set in \mathbf{R}^m , let a mapping $f : \mathbf{R}^n \rightarrow \mathbf{R}^m$ be C^k in U and such that $f(x) \in V$ for $x \in U$, and, finally, let a mapping $g : \mathbf{R}^m \rightarrow \mathbf{R}^p$ be C^k in V . Then the superposition

$$h(x) = g(f(x))$$

is C^k in U .

Remark A.6.4 For higher order derivatives, in contrast to the first order ones, there is no simple “chain rule” for computing the derivative of superposition. For example, the second-order derivative of the superposition $h(x) = g(f(x))$ of two C^2 -mappings is given by the formula

$$Dh(x)[\Delta x^1, \Delta x^2] = Dg(f(x))[D^2 f(x)[\Delta x^1, \Delta x^2]] + D^2 g(x)[Df(x)[\Delta x^1], Df(x)[\Delta x^2]]$$

(check it!). We see that both the first- and the second-order derivatives of f and g contribute to the second-order derivative of the superposition h .

The only case when there does exist a simple formula for high order derivatives of a superposition is the case when the inner function is affine: if $f(x) = Ax + b$ and $h(x) = g(f(x)) = g(Ax + b)$ with a C^ℓ mapping g , then

$$D^\ell h(x)[\Delta x^1, \dots, \Delta x^\ell] = D^\ell g(Ax + b)[A\Delta x^1, \dots, A\Delta x^\ell]. \quad (\text{A.6.11})$$

A.6.9 Examples of higher-order derivatives

Example 1: Second-order derivative of an affine function $f(x) = a + b^T x$ is, of course, identically zero. Indeed, as we have seen,

$$Df(x)[\Delta x^1] = b^T \Delta x^1$$

is independent of x , and therefore the derivative of $Df(x)[\Delta x^1]$ in x , which should give us the second derivative $D^2 f(x)[\Delta x^1, \Delta x^2]$, is zero. Clearly, the third, the fourth, etc., derivatives of an affine function are zero as well.

Example 2: Second-order derivative of a homogeneous quadratic form $f(x) = x^T A x$ (A is a symmetric $n \times n$ matrix). As we have seen,

$$Df(x)[\Delta x^1] = 2x^T A \Delta x^1.$$

Differentiating in x , we get

$$D^2 f(x)[\Delta x^1, \Delta x^2] = \lim_{t \rightarrow +0} t^{-1} [2(x + t\Delta x^2)^T A \Delta x^1 - 2x^T A \Delta x^1] = 2(\Delta x^2)^T A \Delta x^1,$$

so that

$$D^2 f(x)[\Delta x^1, \Delta x^2] = 2(\Delta x^2)^T A \Delta x^1$$

Note that the second derivative of a quadratic form is independent of x ; consequently, the third, the fourth, etc., derivatives of a quadratic form are identically zero.

Example 3: Second-order derivative of the log-det barrier $F(X) = \ln \text{Det}(X)$. As we have seen, this function of an $n \times n$ matrix is well-defined and differentiable on the set U of matrices with positive determinant (which is an open set in the space $\mathbf{R}^{n \times n}$ of $n \times n$ matrices). In fact, this function is C^∞ in U . Let us compute its second-order derivative. As we remember,

$$DF(X)[\Delta X^1] = \text{Tr}(X^{-1} \Delta X^1). \quad (\text{A.6.12})$$

To differentiate the right hand side in X , let us first find the derivative of the mapping $G(X) = X^{-1}$ which is defined on the open set of non-degenerate $n \times n$ matrices. We have

$$\begin{aligned} DG(X)[\Delta X] &= \lim_{t \rightarrow +0} t^{-1} [(X + t\Delta X)^{-1} - X^{-1}] \\ &= \lim_{t \rightarrow +0} t^{-1} [(X(I + tX^{-1}\Delta X))^{-1} - X^{-1}] \\ &= \lim_{t \rightarrow +0} t^{-1} [(I + t\underbrace{X^{-1}\Delta X}_Y)^{-1} X^{-1} - X^{-1}] \\ &= \left[\lim_{t \rightarrow +0} t^{-1} [(I + tY)^{-1} - I] \right] X^{-1} \\ &= \left[\lim_{t \rightarrow +0} t^{-1} [I - (I + tY)] (I + tY)^{-1} \right] X^{-1} \\ &= \left[\lim_{t \rightarrow +0} [-Y(I + tY)^{-1}] \right] X^{-1} \\ &= -YX^{-1} \\ &= -X^{-1}\Delta XX^{-1} \end{aligned}$$

and we arrive at the important by its own right relation

$$D(X^{-1})[\Delta X] = -X^{-1}\Delta XX^{-1}, \quad [X \in \mathbf{R}^{n \times n}, \text{Det}(X) \neq 0]$$

which is the “matrix extension” of the standard relation $(x^{-1})' = -x^{-2}$, $x \in \mathbf{R}$.

Now we are ready to compute the second derivative of the log-det barrier:

$$\begin{aligned}
F(X) &= \ln \text{Det}(X) \\
&\Downarrow \\
DF(X)[\Delta X^1] &= \text{Tr}(X^{-1} \Delta X^1) \\
&\Downarrow \\
D^2 F(X)[\Delta X^1, \Delta X^2] &= \lim_{t \rightarrow +0} t^{-1} [\text{Tr}((X + t \Delta X^2)^{-1} \Delta X^1) - \text{Tr}(X^{-1} \Delta X^1)] \\
&= \lim_{t \rightarrow +0} \text{Tr}(t^{-1} (X + t \Delta X^2)^{-1} \Delta X^1 - X^{-1} \Delta X^1) \\
&= \lim_{t \rightarrow +0} \text{Tr}([t^{-1} (X + t \Delta X^2)^{-1} - X^{-1}] \Delta X^1) \\
&= \text{Tr}([-X^{-1} \Delta X^2 X^{-1}] \Delta X^1),
\end{aligned}$$

and we arrive at the formula

$$D^2 F(X)[\Delta X^1, \Delta X^2] = -\text{Tr}(X^{-1} \Delta X^2 X^{-1} \Delta X^1) \quad [X \in \mathbf{R}^{n \times n}, \text{Det}(X) > 0]$$

Since $\text{Tr}(AB) = \text{Tr}(BA)$ (check it!) for all matrices A, B such that the product AB makes sense and is square, the right hand side in the above formula is symmetric in $\Delta X^1, \Delta X^2$, as it should be for the second derivative of a C^2 function.

A.6.10 Taylor expansion

Assume that $f : \mathbf{R}^n \rightarrow \mathbf{R}^m$ is C^k in a neighbourhood U of a point \bar{x} . The *Taylor expansion of order k* of f , built at the point \bar{x} , is the function

$$\begin{aligned}
F_k(x) &= f(\bar{x}) + \frac{1}{1!} Df(\bar{x})[x - \bar{x}] + \frac{1}{2!} D^2 f(\bar{x})[x - \bar{x}, x - \bar{x}] \\
&\quad + \frac{1}{3!} D^3 f(\bar{x})[x - \bar{x}, x - \bar{x}, x - \bar{x}] + \dots + \frac{1}{k!} D^k f(\bar{x})[\underbrace{x - \bar{x}, \dots, x - \bar{x}}_{k \text{ times}}]
\end{aligned} \tag{A.6.13}$$

We are already acquainted with the Taylor expansion of order 1

$$F_1(x) = f(\bar{x}) + Df(\bar{x})[x - \bar{x}]$$

– this is the affine function of x which approximates “very well” $f(x)$ in a neighbourhood of \bar{x} , namely, within approximation error $\bar{o}(|x - \bar{x}|)$. Similar fact is true for Taylor expansions of higher order:

Theorem A.6.4 *Let $f : \mathbf{R}^n \rightarrow \mathbf{R}^m$ be C^k in a neighbourhood of \bar{x} , and let $F_k(x)$ be the Taylor expansion of f at \bar{x} of degree k . Then*

(i) *$F_k(x)$ is a vector-valued polynomial of full degree $\leq k$ (i.e., every one of the coordinates of the vector $F_k(x)$ is a polynomial of x_1, \dots, x_n , and the sum of powers of x_i 's in every term of this polynomial does not exceed k);*

(ii) *$F_k(x)$ approximates $f(x)$ in a neighbourhood of \bar{x} up to a remainder which is $\bar{o}(|x - \bar{x}|^k)$ as $x \rightarrow \bar{x}$:*

For every $\epsilon > 0$, there exists $\delta > 0$ such that

$$|x - \bar{x}| \leq \delta \Rightarrow |F_k(x) - f(x)| \leq \epsilon |x - \bar{x}|^k.$$

$F_k(\cdot)$ is the unique polynomial with components of full degree $\leq k$ which approximates f up to a remainder which is $\bar{o}(|x - \bar{x}|^k)$.

(iii) The value and the derivatives of F_k of orders $1, 2, \dots, k$, taken at \bar{x} , are the same as the value and the corresponding derivatives of f taken at the same point.

As stated in Theorem, $F_k(x)$ approximates $f(x)$ for x close to \bar{x} up to a remainder which is $\bar{o}(|x - \bar{x}|^k)$. In many cases, it is not enough to know that the remainder is “ $\bar{o}(|x - \bar{x}|^k)$ ” — we need an explicit bound on this remainder. The standard bound of this type is as follows:

Theorem A.6.5 Let k be a positive integer, and let $f : \mathbf{R}^n \rightarrow \mathbf{R}^m$ be C^{k+1} in a ball $B_r = B_r(\bar{x}) = \{x \in \mathbf{R}^n : |x - \bar{x}| < r\}$ of a radius $r > 0$ centered at a point \bar{x} . Assume that the directional derivatives of order $k + 1$, taken at every point of B_r along every unit direction, do not exceed certain $L < \infty$:

$$|D^{k+1}f(x)[d, \dots, d]| \leq L \quad \forall(x \in B_r) \forall(d, |d| = 1).$$

Then for the Taylor expansion F_k of order k of f taken at \bar{x} one has

$$|f(x) - F_k(x)| \leq \frac{L|x - \bar{x}|^{k+1}}{(k+1)!} \quad \forall(x \in B_r).$$

Thus, in a neighbourhood of \bar{x} the remainder of the k -th order Taylor expansion, taken at \bar{x} , is of order of $L|x - \bar{x}|^{k+1}$, where L is the maximal (over all unit directions and all points from the neighbourhood) magnitude of the directional derivatives of order $k + 1$ of f .

A.7 Symmetric matrices

A.7.1 Spaces of matrices

Let \mathbf{S}^m be the space of symmetric $m \times m$ matrices, and $\mathbf{M}^{m,n}$ be the space of rectangular $m \times n$ matrices with real entries. From the viewpoint of their linear structure (i.e., the operations of addition and multiplication by reals) \mathbf{S}^m is just the arithmetic linear space $\mathbf{R}^{m(m+1)/2}$ of dimension $\frac{m(m+1)}{2}$: by arranging the elements of a symmetric $m \times m$ matrix X in a single column, say, in the row-by-row order, you get a usual m^2 -dimensional column vector; multiplication of a matrix by a real and addition of matrices correspond to the same operations with the “representing vector(s)”. When X runs through \mathbf{S}^m , the vector representing X runs through $m(m+1)/2$ -dimensional subspace of \mathbf{R}^{m^2} consisting of vectors satisfying the “symmetry condition” – the coordinates coming from symmetric to each other pairs of entries in X are equal to each other. Similarly, $\mathbf{M}^{m,n}$ as a linear space is just \mathbf{R}^{mn} , and it is natural to equip $\mathbf{M}^{m,n}$ with the inner product defined as the usual inner product of the vectors representing the matrices:

$$\langle X, Y \rangle = \sum_{i=1}^m \sum_{j=1}^n X_{ij} Y_{ij} = \text{Tr}(X^T Y).$$

Here Tr stands for the *trace* – the sum of diagonal elements of a (square) matrix. With this inner product (called the *Frobenius inner product*), $\mathbf{M}^{m,n}$ becomes a legitimate Euclidean space, and we may use in connection with this space all notions based upon the Euclidean structure, e.g., the (Frobenius) norm of a matrix

$$\|X\|_2 = \sqrt{\langle X, X \rangle} = \sqrt{\sum_{i=1}^m \sum_{j=1}^n X_{ij}^2} = \sqrt{\text{Tr}(X^T X)}$$

and likewise the notions of orthogonality, orthogonal complement of a linear subspace, etc. The same applies to the space \mathbf{S}^m equipped with the Frobenius inner product; of course, the Frobenius inner product of symmetric matrices can be written without the transposition sign:

$$\langle X, Y \rangle = \text{Tr}(XY), \quad X, Y \in \mathbf{S}^m.$$

A.7.2 Main facts on symmetric matrices

Let us focus on the space \mathbf{S}^m of symmetric matrices. The most important property of these matrices is as follows:

Theorem A.7.1 [Eigenvalue decomposition] *$n \times n$ matrix A is symmetric if and only if it admits an orthonormal system of eigenvectors: there exist orthonormal basis $\{e_1, \dots, e_n\}$ such that*

$$Ae_i = \lambda_i e_i, \quad i = 1, \dots, n, \tag{A.7.1}$$

for reals λ_i .

In connection with Theorem A.7.1, it is worthy to recall the following notions and facts:

A.7.2.A. Eigenvectors and eigenvalues. An eigenvector of an $n \times n$ matrix A is a nonzero vector e (real or complex) such that $Ae = \lambda e$ for (real or complex) scalar λ ; this scalar is called the eigenvalue of A corresponding to the eigenvector e .

Eigenvalues of A are exactly the roots of the *characteristic polynomial*

$$\pi(z) = \text{Det}(zI - A) = z^n + b_1 z^{n-1} + b_2 z^{n-2} + \dots + b_n$$

of A .

Theorem A.7.1 states, in particular, that for a symmetric matrix A , all eigenvalues are real, and the corresponding eigenvectors can be chosen to be real and to form an orthonormal basis in \mathbf{R}^n .

A.7.2.B. Eigenvalue decomposition of a symmetric matrix. Theorem A.7.1 admits equivalent reformulation as follows (check the equivalence!):

Theorem A.7.2 *An $n \times n$ matrix A is symmetric if and only if it can be represented in the form*

$$A = U\Lambda U^T, \tag{A.7.2}$$

where

- U is an orthogonal matrix: $U^{-1} = U^T$ (or, which is the same, $U^T U = I$, or, which is the same, $U U^T = I$, or, which is the same, the columns of U form an orthonormal basis in \mathbf{R}^n , or, which is the same, the columns of U form an orthonormal basis in \mathbf{R}^n).
- Λ is the diagonal matrix with the diagonal entries $\lambda_1, \dots, \lambda_n$.

Representation (A.7.2) with orthogonal U and diagonal Λ is called the eigenvalue decomposition of A . In such a representation,

- The columns of U form an orthonormal system of eigenvectors of A ;
- The diagonal entries in Λ are the eigenvalues of A corresponding to these eigenvectors.

A.7.2.C. Vector of eigenvalues. When speaking about eigenvalues $\lambda_i(A)$ of a symmetric $n \times n$ matrix A , we always arrange them in the non-ascending order:

$$\lambda_1(A) \geq \lambda_2(A) \geq \dots \geq \lambda_n(A);$$

$\lambda(A) \in \mathbf{R}^n$ denotes the vector of eigenvalues of A taken in the above order.

A.7.2.D. Freedom in eigenvalue decomposition. Part of the data Λ , U in the eigenvalue decomposition (A.7.2) is uniquely defined by A , while the other data admit certain “freedom”. Specifically, the sequence $\lambda_1, \dots, \lambda_n$ of eigenvalues of A (i.e., diagonal entries of Λ) is exactly the sequence of roots of the characteristic polynomial of A (every root is repeated according to its multiplicity) and thus is uniquely defined by A (provided that we arrange the entries of the sequence in the non-ascending order). The columns of U are not uniquely defined by A . What is uniquely defined, are the *linear spans* $E(\lambda)$ of the columns of U corresponding to all eigenvalues equal to certain λ ; such a linear span is nothing but the *spectral subspace* $\{x : Ax = \lambda x\}$ of A corresponding to the eigenvalue λ . There are as many spectral subspaces as many different eigenvalues; spectral subspaces corresponding to different eigenvalues of symmetric matrix are orthogonal to each other, and their sum is the entire space. When building an orthogonal matrix U in the spectral decomposition, one chooses an orthonormal eigenbasis in the spectral subspace corresponding to the largest eigenvalue and makes the vectors of this basis the first columns in U , then chooses an orthonormal basis in the spectral subspace corresponding to the second largest eigenvalue and makes the vector from this basis the next columns of U , and so on.

A.7.2.E. “Simultaneous” decomposition of commuting symmetric matrices. Let A_1, \dots, A_k be $n \times n$ symmetric matrices. It turns out that the matrices commute with each other ($A_i A_j = A_j A_i$ for all i, j) if and only if they can be “simultaneously diagonalized”, i.e., there exist a single orthogonal matrix U and diagonal matrices $\Lambda_1, \dots, \Lambda_k$ such that

$$A_i = U \Lambda_i U^T, \quad i = 1, \dots, k.$$

You are welcome to prove this statement by yourself; to simplify your task, here are two simple and important by their own right statements which help to reach your target:

A.7.2.E.1: Let λ be a real and A, B be two commuting $n \times n$ matrices. Then the spectral subspace $E = \{x : Ax = \lambda x\}$ of A corresponding to λ is invariant for B (i.e., $Be \in E$ for every $e \in E$).

A.7.2.E.2: If A is an $n \times n$ matrix and L is an invariant subspace of A (i.e., L is a linear subspace such that $Ae \in L$ whenever $e \in L$), then the orthogonal complement L^\perp of L is invariant for the matrix A^T . In particular, if A is symmetric and L is invariant subspace of A , then L^\perp is invariant subspace of A as well.

A.7.3 Variational characterization of eigenvalues

Theorem A.7.3 [VCE – Variational Characterization of Eigenvalues] Let A be a symmetric matrix. Then

$$\lambda_\ell(A) = \min_{E \in \mathcal{E}_\ell} \max_{x \in E, x^T x = 1} x^T A x, \quad \ell = 1, \dots, n, \tag{A.7.3}$$

where \mathcal{E}_ℓ is the family of all linear subspaces in \mathbf{R}^n of the dimension $n - \ell + 1$.

VCE says that to get the largest eigenvalue $\lambda_1(A)$, you should maximize the quadratic form $x^T Ax$ over the unit sphere $S = \{x \in \mathbf{R}^n : x^T x = 1\}$; the maximum is exactly $\lambda_1(A)$. To get the second largest eigenvalue $\lambda_2(A)$, you should act as follows: you choose a linear subspace E of dimension $n - 1$ and maximize the quadratic form $x^T Ax$ over the cross-section of S by this subspace; the maximum value of the form depends on E , and you minimize this maximum over linear subspaces E of the dimension $n - 1$; the result is exactly $\lambda_2(A)$. To get $\lambda_3(A)$, you replace in the latter construction subspaces of the dimension $n - 1$ by those of the dimension $n - 2$, and so on. In particular, the smallest eigenvalue $\lambda_n(A)$ is just the minimum, over all linear subspaces E of the dimension $n - n + 1 = 1$, i.e., over all lines passing through the origin, of the quantities $x^T Ax$, where $x \in E$ is unit ($x^T x = 1$); in other words, $\lambda_n(A)$ is just the minimum of the quadratic form $x^T Ax$ over the unit sphere S .

Proof of the VCE is pretty easy. Let e_1, \dots, e_n be an orthonormal eigenbasis of A : $Ae_\ell = \lambda_\ell(A)e_\ell$. For $1 \leq \ell \leq n$, let $F_\ell = \text{Lin}\{e_1, \dots, e_\ell\}$, $G_\ell = \text{Lin}\{e_\ell, e_{\ell+1}, \dots, e_n\}$. Finally, for $x \in \mathbf{R}^n$ let $\xi(x)$ be the vector of coordinates of x in the orthonormal basis e_1, \dots, e_n . Note that

$$x^T x = \xi^T(x) \xi(x),$$

since $\{e_1, \dots, e_n\}$ is an orthonormal basis, and that

$$\begin{aligned} x^T Ax &= x^T A \sum_i \xi_i(x) e_i = x^T \sum_i \lambda_i(A) \xi_i(x) e_i = \\ &\sum_i \lambda_i(A) \xi_i(x) \underbrace{(x^T e_i)}_{\xi_i(x)} \\ &= \sum_i \lambda_i(A) \xi_i^2(x). \end{aligned} \tag{A.7.4}$$

Now, given ℓ , $1 \leq \ell \leq n$, let us set $E = G_\ell$; note that E is a linear subspace of the dimension $n - \ell + 1$. In view of (A.7.4), the maximum of the quadratic form $x^T Ax$ over the intersection of our E with the unit sphere is

$$\max \left\{ \sum_{i=\ell}^n \lambda_i(A) \xi_i^2 : \sum_{i=\ell}^n \xi_i^2 = 1 \right\},$$

and the latter quantity clearly equals to $\max_{\ell \leq i \leq n} \lambda_i(A) = \lambda_\ell(A)$. Thus, for appropriately chosen $E \in \mathcal{E}_\ell$, the inner maximum in the right hand side of (A.7.3) equals to $\lambda_\ell(A)$, whence the right hand side of (A.7.3) is $\leq \lambda_\ell(A)$. It remains to prove the opposite inequality. To this end, consider a linear subspace E of the dimension $n - \ell + 1$ and observe that it has nontrivial intersection with the linear subspace F_ℓ of the dimension ℓ (indeed, $\dim E + \dim F_\ell = (n - \ell + 1) + \ell > n$, so that $\dim(E \cap F) > 0$ by the Dimension formula). It follows that there exists a unit vector y belonging to both E and F_ℓ . Since y is a unit vector from F_ℓ , we have

$$y = \sum_{i=1}^{\ell} \eta_i e_i \text{ with } \sum_{i=1}^{\ell} \eta_i^2 = 1, \text{ whence, by (A.7.4),}$$

$$y^T A y = \sum_{i=1}^{\ell} \lambda_i(A) \eta_i^2 \geq \min_{1 \leq i \leq \ell} \lambda_i(A) = \lambda_\ell(A).$$

Since y is in E , we conclude that

$$\max_{x \in E: x^T x = 1} x^T Ax \geq y^T A y \geq \lambda_\ell(A).$$

Since E is an arbitrary subspace from \mathcal{E}_ℓ , we conclude that the right hand side in (A.7.3) is $\geq \lambda_\ell(A)$. ■

A simple and useful byproduct of our reasoning is the relation (A.7.4):

Corollary A.7.1 *For a symmetric matrix A , the quadratic form $x^T Ax$ is weighted sum of squares of the coordinates $\xi_i(x)$ of x taken with respect to an orthonormal eigenbasis of A ; the weights in this sum are exactly the eigenvalues of A :*

$$x^T Ax = \sum_i \lambda_i(A) \xi_i^2(x).$$

A.7.3.1 Corollaries of the VCE

VCE admits a number of extremely important corollaries as follows:

A.7.3.A. Eigenvalue characterization of positive (semi)definite matrices. Recall that a matrix A is called positive definite (notation: $A \succ 0$), if it is symmetric and the quadratic form $x^T Ax$ is positive outside the origin; A is called positive semidefinite (notation: $A \succeq 0$), if A is symmetric and the quadratic form $x^T Ax$ is nonnegative everywhere. VCE provides us with the following eigenvalue characterization of positive (semi)definite matrices:

Proposition A.7.1 : *A symmetric matrix A is positive semidefinite if and only if its eigenvalues are nonnegative; A is positive definite if and only if all eigenvalues of A are positive*

Indeed, A is positive definite, if and only if the minimum value of $x^T Ax$ over the unit sphere is positive, and is positive semidefinite, if and only if this minimum value is nonnegative; it remains to note that by VCE, the minimum value of $x^T Ax$ over the unit sphere is exactly the minimum eigenvalue of A .

A.7.3.B. \succeq -Monotonicity of the vector of eigenvalues. Let us write $A \succeq B$ ($A \succ B$) to express that A, B are symmetric matrices of the same size such that $A - B$ is positive semidefinite (respectively, positive definite).

Proposition A.7.2 *If $A \succeq B$, then $\lambda(A) \geq \lambda(B)$, and if $A \succ B$, then $\lambda(A) > \lambda(B)$.*

Indeed, when $A \succeq B$, then, of course,

$$\max_{x \in E: x^T x = 1} x^T Ax \geq \max_{x \in E: x^T x = 1} x^T Bx$$

for every linear subspace E , whence

$$\lambda_\ell(A) = \min_{E \in \mathcal{E}_\ell} \max_{x \in E: x^T x = 1} x^T Ax \geq \min_{E \in \mathcal{E}_\ell} \max_{x \in E: x^T x = 1} x^T Bx = \lambda_\ell(B), \quad \ell = 1, \dots, n,$$

i.e., $\lambda(A) \geq \lambda(B)$. The case of $A \succ B$ can be considered similarly.

A.7.3.C. Eigenvalue Interlacement Theorem. We shall formulate this extremely important theorem as follows:

Theorem A.7.4 [Eigenvalue Interlacement Theorem] *Let A be a symmetric $n \times n$ matrix and \bar{A} be the angular $(n-k) \times (n-k)$ submatrix of A . Then, for every $\ell \leq n-k$, the ℓ -th eigenvalue of \bar{A} separates the ℓ -th and the $(\ell+k)$ -th eigenvalues of A :*

$$\lambda_\ell(A) \succeq \lambda_\ell(\bar{A}) \succeq \lambda_{\ell+k}(A). \tag{A.7.5}$$

Indeed, by VCE, $\lambda_\ell(\bar{A}) = \min_{E \in \bar{\mathcal{E}}_\ell} \max_{x \in E: x^T x = 1} x^T \bar{A} x$, where $\bar{\mathcal{E}}_\ell$ is the family of all linear subspaces of the dimension $n - k - \ell + 1$ contained in the linear subspace $\{x \in \mathbf{R}^n : x_{n-k+1} = x_{n-k+2} = \dots = x_n = 0\}$. Since $\bar{\mathcal{E}}_\ell \subset \mathcal{E}_{\ell+k}$, we have

$$\lambda_\ell(\bar{A}) = \min_{E \in \bar{\mathcal{E}}_\ell} \max_{x \in E: x^T x = 1} x^T \bar{A} x \geq \min_{E \in \mathcal{E}_{\ell+k}} \max_{x \in E: x^T x = 1} x^T \bar{A} x = \lambda_{\ell+k}(A).$$

We have proved the left inequality in (A.7.5). Applying this inequality to the matrix $-A$, we get

$$-\lambda_\ell(\bar{A}) = \lambda_{n-k-\ell}(-\bar{A}) \geq \lambda_{n-\ell}(-A) = -\lambda_\ell(A),$$

or, which is the same, $\lambda_\ell(\bar{A}) \leq \lambda_\ell(A)$, which is the first inequality in (A.7.5).

A.7.4 Positive semidefinite matrices and the semidefinite cone

A.7.4.A. Positive semidefinite matrices. Recall that an $n \times n$ matrix A is called *positive semidefinite* (notation: $A \succeq 0$), if A is symmetric and produces nonnegative quadratic form:

$$A \succeq 0 \Leftrightarrow \{A = A^T \text{ and } x^T A x \geq 0 \quad \forall x\}.$$

A is called positive *definite* (notation: $A \succ 0$), if it is positive semidefinite and the corresponding quadratic form is positive outside the origin:

$$A \succ 0 \Leftrightarrow \{A = A^T \text{ and } x^T A x > 0 \quad \forall x \neq 0\}.$$

It makes sense to list a number of equivalent definitions of a positive semidefinite matrix:

Theorem A.7.5 *Let A be a symmetric $n \times n$ matrix. Then the following properties of A are equivalent to each other:*

- (i) $A \succeq 0$
- (ii) $\lambda(A) \geq 0$
- (iii) $A = D^T D$ for certain rectangular matrix D
- (iv) $A = \Delta^T \Delta$ for certain upper triangular $n \times n$ matrix Δ
- (v) $A = B^2$ for certain symmetric matrix B ;
- (vi) $A = B^2$ for certain $B \succeq 0$.

The following properties of a symmetric matrix A also are equivalent to each other:

- (i') $A \succ 0$
- (ii') $\lambda(A) > 0$
- (iii') $A = D^T D$ for certain rectangular matrix D of rank n
- (iv') $A = \Delta^T \Delta$ for certain nondegenerate upper triangular $n \times n$ matrix Δ
- (v') $A = B^2$ for certain nondegenerate symmetric matrix B ;
- (vi') $A = B^2$ for certain $B \succ 0$.

Proof. (i) \Leftrightarrow (ii): this equivalence is stated by Proposition A.7.1.

(ii) \Leftrightarrow (vi): Let $A = U \Lambda U^T$ be the eigenvalue decomposition of A , so that U is orthogonal and Λ is diagonal with nonnegative diagonal entries $\lambda_i(A)$ (we are in the situation of (ii)!). Let $\Lambda^{1/2}$ be the diagonal matrix with the diagonal entries $\lambda_i^{1/2}(A)$; note that $(\Lambda^{1/2})^2 = \Lambda$. The matrix $B = U \Lambda^{1/2} U^T$ is symmetric with nonnegative eigenvalues $\lambda_i^{1/2}(A)$, so that $B \succeq 0$ by Proposition A.7.1, and

$$B^2 = U \Lambda^{1/2} \underbrace{U^T U}_{I} \Lambda^{1/2} U^T = U (\Lambda^{1/2})^2 U^T = U \Lambda U^T = A,$$

as required in (vi).

(vi) \Rightarrow (v): evident.

(v) \Rightarrow (iv): Let $A = B^2$ with certain symmetric B , and let b_i be i -th column of B . Applying the Gram-Schmidt orthogonalization process (see proof of Theorem A.2.3.(iii)), we can find an orthonormal system of vectors u_1, \dots, u_n and lower triangular matrix L such that $b_i = \sum_{j=1}^i L_{ij}u_j$, or, which is the same, $B^T = LU$, where U is the orthogonal matrix with the rows u_1^T, \dots, u_n^T . We now have $A = B^2 = B^T(B^T)^T = LUU^T L^T = LL^T$. We see that $A = \Delta^T \Delta$, where the matrix $\Delta = L^T$ is upper triangular.

(iv) \Rightarrow (iii): evident.

(iii) \Rightarrow (i): If $A = D^T D$, then $x^T Ax = (Dx)^T(Dx) \geq 0$ for all x .

We have proved the equivalence of the properties (i) – (vi). Slightly modifying the reasoning (do it yourself!), one can prove the equivalence of the properties (i') – (vi'). ■

Remark A.7.1 (i) [Checking positive semidefiniteness] Given an $n \times n$ symmetric matrix A , one can check whether it is positive semidefinite by a purely algebraic finite algorithm (the so called *Lagrange diagonalization of a quadratic form*) which requires at most $O(n^3)$ arithmetic operations. Positive definiteness of a matrix can be checked also by the *Choleski factorization algorithm* which finds the decomposition in (iv'), if it exists, in approximately $\frac{1}{6}n^3$ arithmetic operations.

There exists another useful algebraic criterion (Sylvester's criterion) for positive semidefiniteness of a matrix; according to this criterion, a symmetric matrix A is positive definite if and only if its angular minors are positive, and A is positive semidefinite if and only if all its principal minors are nonnegative. For example, a symmetric 2×2 matrix $A = \begin{bmatrix} a & b \\ b & c \end{bmatrix}$ is positive semidefinite if and only if $a \geq 0$, $c \geq 0$ and $\text{Det}(A) \equiv ac - b^2 \geq 0$.

(ii) [Square root of a positive semidefinite matrix] By the first chain of equivalences in Theorem A.7.5, a symmetric matrix A is $\succeq 0$ if and only if A is the square of a positive semidefinite matrix B . The latter matrix is uniquely defined by $A \succeq 0$ and is called the *square root* of A (notation: $A^{1/2}$).

A.7.4.B. The semidefinite cone. When adding symmetric matrices and multiplying them by reals, we add, respectively multiply by reals, the corresponding quadratic forms. It follows that

A.7.4.B.1: *The sum of positive semidefinite matrices and a product of a positive semidefinite matrix and a nonnegative real is positive semidefinite,*

or, which is the same (see Section B.1.4),

A.7.4.B.2: *$n \times n$ positive semidefinite matrices form a cone \mathbf{S}_+^n in the Euclidean space \mathbf{S}^n of symmetric $n \times n$ matrices, the Euclidean structure being given by the Frobenius inner product $\langle A, B \rangle = \text{Tr}(AB) = \sum_{i,j} A_{ij}B_{ij}$.*

The cone \mathbf{S}_+^n is called the *semidefinite cone* of size n . It is immediately seen that the semidefinite cone \mathbf{S}_+^n is “good” (see Lecture 5), specifically,

- \mathbf{S}_+^n is closed: the limit of a converging sequence of positive semidefinite matrices is positive semidefinite;

- \mathbf{S}_+^n is pointed: the only $n \times n$ matrix A such that both A and $-A$ are positive semidefinite is the zero $n \times n$ matrix;
- \mathbf{S}_+^n possesses a nonempty interior which is comprised of positive definite matrices.

Note that the relation $A \succeq B$ means exactly that $A - B \in \mathbf{S}_+^n$, while $A \succ B$ is equivalent to $A - B \in \text{int } \mathbf{S}_+^n$. The “matrix inequalities” $A \succeq B$ ($A \succ B$) match the standard properties of the usual scalar inequalities, e.g.:

$$\begin{array}{lll} A \succeq A & & [\text{reflexivity}] \\ A \succeq B, B \succeq A \Rightarrow A = B & & [\text{antisymmetry}] \\ A \succeq B, B \succeq C \Rightarrow A \succeq C & & [\text{transitivity}] \\ A \succeq B, C \succeq D \Rightarrow A + C \succeq B + D & & [\text{compatibility with linear operations, I}] \\ A \succeq B, \lambda \geq 0 \Rightarrow \lambda A \succeq \lambda B & & [\text{compatibility with linear operations, II}] \\ A_i \succeq B_i, A_i \rightarrow A, B_i \rightarrow B \text{ as } i \rightarrow \infty \Rightarrow A \succeq B & & [\text{closedness}] \end{array}$$

with evident modifications when \succeq is replaced with \succ , or

$$A \succeq B, C \succ D \Rightarrow A + C \succ B + D,$$

etc. Along with these standard properties of inequalities, the inequality \succeq possesses a nice additional property:

A.7.4.B.3: *In a valid \succeq -inequality*

$$A \succeq B$$

one can multiply both sides from the left and by the right by a (rectangular) matrix and its transpose:

$$\begin{aligned} A, B \in \mathbf{S}^n, \quad A \succeq B, \quad V \in \mathbf{M}^{n,m} \\ \Downarrow \\ V^T A V \succeq V^T B V \end{aligned}$$

Indeed, we should prove that if $A - B \succeq 0$, then also $V^T(A - B)V \succeq 0$, which is immediate – the quadratic form $y^T[V^T(A - B)V]y = (Vy)^T(A - B)(Vy)$ of y is nonnegative along with the quadratic form $x^T(A - B)x$ of x .

An important additional property of the semidefinite cone is its *self-duality*:

Theorem A.7.6 *A symmetric matrix Y has nonnegative Frobenius inner products with all positive semidefinite matrices if and only if Y itself is positive semidefinite.*

Proof. “if” part: Assume that $Y \succeq 0$, and let us prove that then $\text{Tr}(YX) \geq 0$ for every $X \succeq 0$. Indeed, the eigenvalue decomposition of Y can be written as

$$Y = \sum_{i=1}^n \lambda_i(Y) e_i e_i^T,$$

where e_i are the orthonormal eigenvectors of Y . We now have

$$\begin{aligned} \text{Tr}(YX) &= \text{Tr}\left(\left(\sum_{i=1}^n \lambda_i(Y) e_i e_i^T\right) X\right) = \sum_{i=1}^n \lambda_i(Y) \text{Tr}(e_i e_i^T X) \\ &= \sum_{i=1}^n \lambda_i(Y) \text{Tr}(e_i^T X e_i), \end{aligned} \tag{A.7.6}$$

where the concluding equality is given by the following well-known property of the trace:

A.7.4.B.4: Whenever matrices A, B are such that the product AB makes sense and is a square matrix, one has

$$\text{Tr}(AB) = \text{Tr}(BA).$$

Indeed, we should verify that if $A \in \mathbf{M}^{p,q}$ and $B \in \mathbf{M}^{q,p}$, then $\text{Tr}(AB) = \text{Tr}(BA)$. The left hand side quantity in our hypothetic equality is $\sum_{i=1}^p \sum_{j=1}^q A_{ij}B_{ji}$, and the right hand side quantity is $\sum_{j=1}^q \sum_{i=1}^p B_{ji}A_{ij}$; they indeed are equal.

Looking at the concluding quantity in (A.7.6), we see that it indeed is nonnegative whenever $X \succeq 0$ (since $Y \succeq 0$ and thus $\lambda_i(Y) \geq 0$ by P.7.5).

"only if" part: We are given Y such that $\text{Tr}(YX) \geq 0$ for all matrices $X \succeq 0$, and we should prove that $Y \succeq 0$. This is immediate: for every vector x , the matrix $X = xx^T$ is positive semidefinite (Theorem A.7.5.(iii)), so that $0 \leq \text{Tr}(Yxx^T) = \text{Tr}(x^TYx) = x^TYx$. Since the resulting inequality $x^TYx \geq 0$ is valid for every x , we have $Y \succeq 0$. ■

Appendix B

Convex sets in \mathbf{R}^n

B.1 Definition and basic properties

B.1.1 A convex set

In the school geometry a figure is called convex if it contains, along with every pair of its points x, y , also the entire segment $[x, y]$ linking the points. This is exactly the definition of a convex set in the multidimensional case; all we need is to say what does it mean “the segment $[x, y]$ linking the points $x, y \in \mathbf{R}^n$ ”. This is said by the following

Definition B.1.1 [Convex set]

1) Let x, y be two points in \mathbf{R}^n . The set

$$[x, y] = \{z = \lambda x + (1 - \lambda)y : 0 \leq \lambda \leq 1\}$$

is called a segment with the endpoints x, y .

2) A subset M of \mathbf{R}^n is called convex, if it contains, along with every pair of its points x, y , also the entire segment $[x, y]$:

$$x, y \in M, 0 \leq \lambda \leq 1 \Rightarrow \lambda x + (1 - \lambda)y \in M.$$

Note that by this definition an empty set is convex (by convention, or better to say, by the exact sense of the definition: for the empty set, you cannot present a counterexample to show that it is not convex).

B.1.2 Examples of convex sets

B.1.2.A. Affine subspaces and polyhedral sets

Example B.1.1 A linear/affine subspace of \mathbf{R}^n is convex.

Convexity of affine subspaces immediately follows from the possibility to represent these sets as solution sets of systems of linear equations (Proposition A.3.7), due to the following simple and important fact:

Proposition B.1.1 The solution set of an arbitrary (possibly, infinite) system

$$a_\alpha^T x \leq b_\alpha, \alpha \in \mathcal{A} \quad (!)$$

of nonstrict linear inequalities with n unknowns x – the set

$$S = \{x \in \mathbf{R}^n : a_\alpha^T x \leq b_\alpha, \alpha \in \mathcal{A}\}$$

is convex.

In particular, the solution set of a finite system

$$Ax \leq b$$

of m nonstrict inequalities with n variables (A is $m \times n$ matrix) is convex; a set of this latter type is called polyhedral.

Exercise B.1 Prove Proposition B.1.1.

Remark B.1.1 Note that every set given by Proposition B.1.1 is not only convex, but also closed (why?). In fact, from Separation Theorem (Theorem B.2.9 below) it follows that

Every closed convex set in \mathbf{R}^n is the solution set of a (perhaps, infinite) system of nonstrict linear inequalities.

Remark B.1.2 Note that replacing some of the nonstrict linear inequalities $a_\alpha^T x \leq b_\alpha$ in (!) with their strict versions $a_\alpha^T x < b_\alpha$, we get a system with the solution set which still is convex (why?), but now not necessary is closed.

B.1.2.2 B.1.2.B. Unit balls of norms

Let $\|\cdot\|$ be a norm on \mathbf{R}^n i.e., a real-valued function on \mathbf{R}^n satisfying the three characteristic properties of a norm (Section A.4.1), specifically:

A. [positivity] $\|x\| \geq 0$ for all $x \in \mathbf{R}^n$; $\|x\| = 0$ if and only if $x = 0$;

B. [homogeneity] For $x \in \mathbf{R}^n$ and $\lambda \in \mathbf{R}$, one has

$$\|\lambda x\| = |\lambda| \|x\|;$$

C. [triangle inequality] For all $x, y \in \mathbf{R}^n$ one has

$$\|x + y\| \leq \|x\| + \|y\|.$$

Example B.1.2 The unit ball of the norm $\|\cdot\|$ – the set

$$\{x \in E : \|x\| \leq 1\},$$

same as every other $\|\cdot\|$ -ball

$$\{x : \|x - a\| \leq r\}$$

($a \in \mathbf{R}^n$ and $r \geq 0$ are fixed) is convex.

In particular, Euclidean balls ($\|\cdot\|$ -balls associated with the standard Euclidean norm $\|x\|_2 = \sqrt{x^T x}$) are convex.

The standard examples of norms on \mathbf{R}^n are the ℓ_p -norms

$$\|x\|_p = \begin{cases} \left(\sum_{i=1}^n |x_i|^p \right)^{1/p}, & 1 \leq p < \infty \\ \max_{1 \leq i \leq n} |x_i|, & p = \infty \end{cases}.$$

These indeed are norms (which is not clear in advance). When $p = 2$, we get the usual Euclidean norm; of course, you know how the Euclidean ball looks. When $p = 1$, we get

$$\|x\|_1 = \sum_{i=1}^n |x_i|,$$

and the unit ball is the *hyperoctahedron*

$$V = \{x \in \mathbf{R}^n : \sum_{i=1}^n |x_i| \leq 1\}$$

When $p = \infty$, we get

$$\|x\|_\infty = \max_{1 \leq i \leq n} |x_i|,$$

and the unit ball is the *hypercube*

$$V = \{x \in \mathbf{R}^n : -1 \leq x_i \leq 1, 1 \leq i \leq n\}.$$

Exercise B.2 [†] Prove that unit balls of norms on \mathbf{R}^n are exactly the same as convex sets V in \mathbf{R}^n satisfying the following three properties:

1. V is symmetric w.r.t. the origin: $x \in V \Rightarrow -x \in V$;
2. V is bounded and closed;
3. V contains a neighbourhood of the origin.

A set V satisfying the outlined properties is the unit ball of the norm

$$\|x\| = \inf \left\{ t \geq 0 : t^{-1}x \in V \right\}.$$

Hint: You could find useful to verify and to exploit the following facts:

1. A norm $\|\cdot\|$ on \mathbf{R}^n is Lipschitz continuous with respect to the standard Euclidean distance: there exists $C_{\|\cdot\|} < \infty$ such that $\| \|x\| - \|y\| \| \leq C_{\|\cdot\|} \|x - y\|_2$ for all x, y
2. Vice versa, the Euclidean norm is Lipschitz continuous with respect to a given norm $\|\cdot\|$: there exists $c_{\|\cdot\|} < \infty$ such that $\| \|x\|_2 - \|y\|_2 \| \leq c_{\|\cdot\|} \|x - y\|$ for all x, y

B.1.2.3 B.1.2.C. Ellipsoids

Example B.1.3 [Ellipsoid] Let Q be a $n \times n$ matrix which is symmetric ($Q = Q^T$) and positive definite ($x^T Q x \geq 0$, with \geq being $=$ if and only if $x = 0$). Then, for every nonnegative r , the Q -ellipsoid of radius r centered at a – the set

$$\{x : (x - a)^T Q(x - a) \leq r^2\}$$

is convex.

To see that an ellipsoid $\{x : (x - a)^T Q(x - a) \leq r^2\}$ is convex, note that since Q is positive definite, the matrix $Q^{1/2}$ is well-defined and positive definite. Now, if $\|\cdot\|$ is a norm on \mathbf{R}^n and P is a nonsingular $n \times n$ matrix, the function $\|Px\|$ is a norm along with $\|\cdot\|$ (why?). Thus, the function $\|x\|_Q \equiv \sqrt{x^T Q x} = \|Q^{1/2}x\|_2$ is a norm along with $\|\cdot\|_2$, and the ellipsoid in question clearly is just $\|\cdot\|_Q$ -ball of radius r centered at a .

B.1.2.4 B.1.2.D. Neighbourhood of a convex set

Example B.1.4 Let M be a convex set in \mathbf{R}^n , and let $\epsilon > 0$. Then, for every norm $\|\cdot\|$ on \mathbf{R}^n , the ϵ -neighbourhood of M , i.e., the set

$$M_\epsilon = \{y \in \mathbf{R}^n : \text{dist}_{\|\cdot\|}(y, M) \equiv \inf_{x \in M} \|y - x\| \leq \epsilon\}$$

is convex.

Exercise B.3 Justify the statement of Example B.1.4.

B.1.3 Inner description of convex sets: Convex combinations and convex hull

B.1.3.1 B.1.3.A. Convex combinations

Recall the notion of *linear combination* y of vectors y_1, \dots, y_m – this is a vector represented as

$$y = \sum_{i=1}^m \lambda_i y_i,$$

where λ_i are real coefficients. Specifying this definition, we have come to the notion of an *affine combination* – this is a linear combination with the sum of coefficients equal to one. The last notion in this genre is the one of *convex combination*.

Definition B.1.2 A convex combination of vectors y_1, \dots, y_m is their affine combination with nonnegative coefficients, or, which is the same, a linear combination

$$y = \sum_{i=1}^m \lambda_i y_i$$

with nonnegative coefficients with unit sum:

$$\lambda_i \geq 0, \quad \sum_{i=1}^m \lambda_i = 1.$$

The following statement resembles those in Corollary A.3.2:

Proposition B.1.2 *A set M in \mathbf{R}^n is convex if and only if it is closed with respect to taking all convex combinations of its elements, i.e., if and only if every convex combination of vectors from M again is a vector from M .*

Exercise B.4 *Prove Proposition B.1.2.*

Hint: Assuming $\lambda_1, \dots, \lambda_m > 0$, one has

$$\sum_{i=1}^m \lambda_i y_i = \lambda_1 y_1 + (\lambda_2 + \lambda_3 + \dots + \lambda_m) \sum_{i=2}^m \mu_i y_i, \quad \mu_i = \frac{\lambda_i}{\lambda_2 + \lambda_3 + \dots + \lambda_m}.$$

B.1.3.2 B.1.3.B. Convex hull

Same as the property to be linear/affine subspace, the property to be convex is preserved by taking intersections (why?):

Proposition B.1.3 *Let $\{M_\alpha\}_\alpha$ be an arbitrary family of convex subsets of \mathbf{R}^n . Then the intersection*

$$M = \cap_\alpha M_\alpha$$

is convex.

As an immediate consequence, we come to the notion of *convex hull* $\text{Conv}(M)$ of a nonempty subset in \mathbf{R}^n (cf. the notions of linear/affine hull):

Corollary B.1.1 [Convex hull]

Let M be a nonempty subset in \mathbf{R}^n . Then among all convex sets containing M (these sets do exist, e.g., \mathbf{R}^n itself) there exists the smallest one, namely, the intersection of all convex sets containing M . This set is called the convex hull of M [notation: $\text{Conv}(M)$].

The linear span of M is the set of all linear combinations of vectors from M , the affine hull is the set of all affine combinations of vectors from M . As you guess,

Proposition B.1.4 [Convex hull via convex combinations] *For a nonempty $M \subset \mathbf{R}^n$:*

$$\text{Conv}(M) = \{ \text{the set of all convex combinations of vectors from } M \}.$$

Exercise B.5 *Prove Proposition B.1.4.*

B.1.3.3 B.1.3.C. Simplex

The convex hull of $m + 1$ affinely independent points y_0, \dots, y_m (Section A.3.3) is called *m -dimensional simplex with the vertices y_0, \dots, y_m* . By results of Section A.3.3, every point x of an m -dimensional simplex with vertices y_0, \dots, y_m admits exactly one representation as a convex combination of the vertices; the corresponding coefficients form the unique solution to the system of linear equations

$$\sum_{i=0}^m \lambda_i x_i = x, \quad \sum_{i=0}^m \lambda_i = 1.$$

This system is solvable if and only if $x \in M = \text{Aff}(\{y_0, \dots, y_m\})$, and the components of the solution (the barycentric coordinates of x) are affine functions of $x \in \text{Aff}(M)$; the simplex itself is comprised of points from M with nonnegative barycentric coordinates.

B.1.4 Cones

A nonempty subset M of \mathbf{R}^n is called *conic*, if it contains, along with every point $x \in M$, the entire ray $\mathbf{R}x = \{tx : t \geq 0\}$ spanned by the point:

$$x \in M \Rightarrow tx \in M \quad \forall t \geq 0.$$

A convex conic set is called a *cone*.

Proposition B.1.5 *A nonempty subset M of \mathbf{R}^n is a cone if and only if it possesses the following pair of properties:*

- *is conic:* $x \in M, t \geq 0 \Rightarrow tx \in M$;
- *contains sums of its elements:* $x, y \in M \Rightarrow x + y \in M$.

Exercise B.6 *Prove Proposition B.1.5.*

As an immediate consequence, we get that a cone is closed with respect to taking linear combinations with *nonnegative coefficients* of the elements, and vice versa – a nonempty set closed with respect to taking these combinations is a cone.

Example B.1.5 *The solution set of an arbitrary (possibly, infinite) system*

$$a_\alpha^T x \leq 0, \quad \alpha \in \mathcal{A}$$

of homogeneous linear inequalities with n unknowns x – the set

$$K = \{x : a_\alpha^T x \leq 0 \quad \forall \alpha \in \mathcal{A}\}$$

– *is a cone.*

In particular, the solution set to a homogeneous finite system of m homogeneous linear inequalities

$$Ax \leq 0$$

(A is $m \times n$ matrix) is a cone; a cone of this latter type is called polyhedral.

Note that the cones given by systems of linear homogeneous nonstrict inequalities necessarily are closed. From Separation Theorem B.2.9 it follows that, vice versa, every closed convex cone is the solution set to such a system, so that Example B.1.5 is the generic example of a closed convex cone.

Cones form a very important family of convex sets, and one can develop theory of cones absolutely similar (and in a sense, equivalent) to that one of all convex sets. E.g., introducing the notion of *conic combination* of vectors x_1, \dots, x_k as a linear combination of the vectors with nonnegative coefficients, you can easily prove the following statements completely similar to those for general convex sets, with conic combination playing the role of convex one:

- A set is a cone if and only if it is nonempty and is closed with respect to taking all conic combinations of its elements;
- Intersection of a family of cones is again a cone; in particular, for every nonempty set $M \subset \mathbf{R}^n$ there exists the smallest cone containing M – its conic hull $\text{Cone}(M)$, and this conic hull is comprised of all conic combinations of vectors from M .

In particular, the conic hull of a nonempty finite set $M = \{u_1, \dots, u_N\}$ of vectors in \mathbf{R}^n is the cone

$$\text{Cone}(M) = \left\{ \sum_{i=1}^N \lambda_i u_i : \lambda_i \geq 0, i = 1, \dots, N \right\}.$$

B.1.5 Calculus of convex sets

Proposition B.1.6 *The following operations preserve convexity of sets:*

1. Taking intersection: if M_α , $\alpha \in \mathcal{A}$, are convex sets, so is the set $\bigcap_\alpha M_\alpha$.
 2. Taking direct product: if $M_1 \subset \mathbf{R}^{n_1}$ and $M_2 \subset \mathbf{R}^{n_2}$ are convex sets, so is the set
- $$M_1 \times M_2 = \{y = (y_1, y_2) \in \mathbf{R}^{n_1} \times \mathbf{R}^{n_2} = \mathbf{R}^{n_1+n_2} : y_1 \in M_1, y_2 \in M_2\}.$$
3. Arithmetic summation and multiplication by reals: if M_1, \dots, M_k are convex sets in \mathbf{R}^n and $\lambda_1, \dots, \lambda_k$ are arbitrary reals, then the set

$$\lambda_1 M_1 + \dots + \lambda_k M_k = \left\{ \sum_{i=1}^k \lambda_i x_i : x_i \in M_i, i = 1, \dots, k \right\}$$

is convex.

4. Taking the image under an affine mapping: if $M \subset \mathbf{R}^n$ is convex and $x \mapsto \mathcal{A}(x) \equiv Ax + b$ is an affine mapping from \mathbf{R}^n into \mathbf{R}^m (A is $m \times n$ matrix, b is m -dimensional vector), then the set

$$\mathcal{A}(M) = \{y = \mathcal{A}(x) \equiv Ax + b : x \in M\}$$

is a convex set in \mathbf{R}^m ;

5. Taking the inverse image under affine mapping: if $M \subset \mathbf{R}^n$ is convex and $y \mapsto Ay + b$ is an affine mapping from \mathbf{R}^m to \mathbf{R}^n (A is $n \times m$ matrix, b is n -dimensional vector), then the set

$$\mathcal{A}^{-1}(M) = \{y \in \mathbf{R}^m : \mathcal{A}(y) \in M\}$$

is a convex set in \mathbf{R}^m .

Exercise B.7 Prove Proposition B.1.6.

B.1.6 Topological properties of convex sets

Convex sets and closely related objects - convex functions - play the central role in Optimization. To play this role properly, the convexity alone is insufficient; we need convexity plus closedness.

B.1.6.1 B.1.6.A. The closure

It is clear from definition of a closed set (Section A.4.3) that the intersection of a family of closed sets in \mathbf{R}^n is also closed. From this fact it, as always, follows that for every subset M of \mathbf{R}^n there exists the smallest closed set containing M ; this set is called the *closure* of M and is denoted $\text{cl } M$. In Analysis they prove the following inner description of the closure of a set in a metric space (and, in particular, in \mathbf{R}^n):

The closure of a set $M \subset \mathbf{R}^n$ is exactly the set comprised of the limits of all converging sequences of elements of M .

With this fact in mind, it is easy to prove that, e.g., the closure of the open Euclidean ball

$$\{x : |x - a| < r\} \quad [r > 0]$$

is the closed ball $\{x : \|x - a\|_2 \leq r\}$. Another useful application example is the closure of a set

$$M = \{x : a_\alpha^T x < b_\alpha, \alpha \in \mathcal{A}\}$$

given by strict linear inequalities: *if such a set is nonempty, then its closure is given by the nonstrict versions of the same inequalities:*

$$\text{cl } M = \{x : a_\alpha^T x \leq b_\alpha, \alpha \in \mathcal{A}\}.$$

Nonemptiness of M in the latter example is essential: the set M given by two strict inequalities

$$x < 0, \quad -x < 0$$

in \mathbf{R} clearly is empty, so that its closure also is empty; in contrast to this, applying formally the above rule, we would get wrong answer

$$\text{cl } M = \{x : x \leq 0, x \geq 0\} = \{0\}.$$

B.1.6.2 B.1.6.B. The interior

Let $M \subset \mathbf{R}^n$. We say that a point $x \in M$ is an *interior point* of M , if some neighbourhood of the point is contained in M , i.e., there exists centered at x ball of positive radius which belongs to M :

$$\exists r > 0 \quad B_r(x) \equiv \{y : \|y - x\|_2 \leq r\} \subset M.$$

The set of all interior points of M is called the *interior* of M [notation: $\text{int } M$].

E.g.,

- The interior of an open set is the set itself;
- The interior of the closed ball $\{x : \|x - a\|_2 \leq r\}$ is the open ball $\{x : \|x - a\|_2 < r\}$ (why?)
- The interior of a polyhedral set $\{x : Ax \leq b\}$ with matrix A not containing zero rows is the set $\{x : Ax < b\}$ (why?)

The latter statement is not, generally speaking, valid for sets of solutions of infinite systems of linear inequalities. E.g., the system of inequalities

$$x \leq \frac{1}{n}, \quad n = 1, 2, \dots$$

in \mathbf{R} has, as a solution set, the nonpositive ray $\mathbf{R}_- = \{x \leq 0\}$; the interior of this ray is the negative ray $\{x < 0\}$. At the same time, strict versions of our inequalities

$$x < \frac{1}{n}, \quad n = 1, 2, \dots$$

define the same nonpositive ray, not the negative one.

It is also easily seen (this fact is valid for arbitrary metric spaces, not for \mathbf{R}^n only), that

- the interior of an arbitrary set is open

The interior of a set is, of course, contained in the set, which, in turn, is contained in its closure:

$$\text{int } M \subset M \subset \text{cl } M. \quad (\text{B.1.1})$$

The complement of the interior in the closure – the set

$$\partial M = \text{cl } M \setminus \text{int } M$$

– is called the *boundary* of M , and the points of the boundary are called *boundary points* of M (Warning: these points not necessarily belong to M , since M can be less than $\text{cl } M$; in fact, all boundary points belong to M if and only if $M = \text{cl } M$, i.e., if and only if M is closed).

The boundary of a set clearly is closed (as the intersection of two closed sets $\text{cl } M$ and $\mathbf{R}^n \setminus \text{int } M$; the latter set is closed as a complement to an open set). From the definition of the boundary,

$$M \subset \text{int } M \cup \partial M \quad [= \text{cl } M],$$

so that a point from M is either an interior, or a boundary point of M .

B.1.6.3 B.1.6.C. The relative interior

Many of the constructions in Optimization possess nice properties in the interior of the set the construction is related to and may lose these nice properties at the boundary points of the set; this is why in many cases we are especially interested in interior points of sets and want the set of these points to be “enough massive”. What to do if it is not the case – e.g., there are no interior points at all (look at a segment in the plane)? It turns out that in these cases we can use a good surrogate of the “normal” interior – the *relative interior* defined as follows.

Definition B.1.3 [Relative interior] Let $M \subset \mathbf{R}^n$. We say that a point $x \in M$ is *relative interior* for M , if M contains the intersection of a small enough ball centered at x with $\text{Aff}(M)$:

$$\exists r > 0 \quad B_r(x) \cap \text{Aff}(M) \equiv \{y : y \in \text{Aff}(M), \|y - x\|_2 \leq r\} \subset M.$$

The set of all relative interior points of M is called its *relative interior* [notation: $\text{ri } M$].

E.g. the relative interior of a singleton is the singleton itself (since a point in the 0-dimensional space is the same as a ball of a positive radius); more generally, the relative interior of an affine subspace is the set itself. The interior of a segment $[x, y]$ ($x \neq y$) in \mathbf{R}^n is empty whenever $n > 1$; in contrast to this, the relative interior is nonempty independently of n and is the interval (x, y) – the segment with deleted endpoints. Geometrically speaking, the relative interior is the interior we get when regard M as a subset of its affine hull (the latter, geometrically, is nothing but \mathbf{R}^k , k being the affine dimension of $\text{Aff}(M)$).

Exercise B.8 Prove that the relative interior of a simplex with vertices y_0, \dots, y_m is exactly the set $\{x = \sum_{i=0}^m \lambda_i y_i : \lambda_i > 0, \sum_{i=0}^m \lambda_i = 1\}$.

We can play with the notion of the relative interior in basically the same way as with the one of interior, namely:

- since $\text{Aff}(M)$, as every affine subspace, is closed and contains M , it contains also the smallest closed sets containing M , i.e., $\text{cl } M$. Therefore we have the following analogies of inclusions (B.1.1):

$$\text{ri } M \subset M \subset \text{cl } M \quad [\subset \text{Aff}(M)]; \quad (\text{B.1.2})$$

- we can define the *relative boundary* $\partial_{\text{ri}} M = \text{cl } M \setminus \text{ri } M$ which is a closed set contained in $\text{Aff}(M)$, and, as for the “actual” interior and boundary, we have

$$\text{ri } M \subset M \subset \text{cl } M = \text{ri } M + \partial_{\text{ri}} M.$$

Of course, if $\text{Aff}(M) = \mathbf{R}^n$, then the relative interior becomes the usual interior, and similarly for boundary; this for sure is the case when $\text{int } M \neq \emptyset$ (since then M contains a ball B , and therefore the affine hull of M is the entire \mathbf{R}^n , which is the affine hull of B).

B.1.6.4 B.1.6.D. Nice topological properties of convex sets

An arbitrary set M in \mathbf{R}^n may possess very pathological topology: both inclusions in the chain

$$\text{ri } M \subset M \subset \text{cl } M$$

can be very “non-tight”. E.g., let M be the set of rational numbers in the segment $[0, 1] \subset \mathbf{R}$. Then $\text{ri } M = \text{int } M = \emptyset$ – since every neighbourhood of every rational real contains irrational reals – while $\text{cl } M = [0, 1]$. Thus, $\text{ri } M$ is “incomparably smaller” than M , $\text{cl } M$ is “incomparably larger”, and M is contained in its relative boundary (by the way, what is this relative boundary?).

The following proposition demonstrates that the topology of a *convex* set M is much better than it might be for an arbitrary set.

Theorem B.1.1 *Let M be a convex set in \mathbf{R}^n . Then*

- The interior $\text{int } M$, the closure $\text{cl } M$ and the relative interior $\text{ri } M$ are convex;*
- If M is nonempty, then the relative interior $\text{ri } M$ of M is nonempty*
- The closure of M is the same as the closure of its relative interior:*

$$\text{cl } M = \text{cl } \text{ri } M$$

(in particular, every point of $\text{cl } M$ is the limit of a sequence of points from $\text{ri } M$)

- The relative interior remains unchanged when we replace M with its closure:*

$$\text{ri } M = \text{ri } \text{cl } M.$$

Proof. (i): prove yourself!

(ii): Let M be a nonempty convex set, and let us prove that $\text{ri } M \neq \emptyset$. By translation, we may assume that $0 \in M$. Further, we may assume that the linear span of M is the entire \mathbf{R}^n . Indeed, as far as linear operations and the Euclidean structure are concerned, the linear span L of M , as every other linear subspace in \mathbf{R}^n , is equivalent to certain \mathbf{R}^k ; since the notion of relative interior deals only with linear and Euclidean structures, we lose nothing thinking of $\text{Lin}(M)$ as of \mathbf{R}^k and taking it as our universe instead of the original universe \mathbf{R}^n . Thus, in the rest of the proof of (ii) we assume that $0 \in M$ and $\text{Lin}(M) = \mathbf{R}^n$; what we should prove is that the interior of M (which in the case in question is the same as relative interior) is nonempty. Note that since $0 \in M$, we have $\text{Aff}(M) = \text{Lin}(M) = \mathbf{R}^n$.

Since $\text{Lin}(M) = \mathbf{R}^n$, we can find in M n linearly independent vectors a_1, \dots, a_n . Let also $a_0 = 0$. The $n+1$ vectors a_0, \dots, a_n belong to M , and since M is convex, the convex hull of these vectors also belongs to M . This convex hull is the set

$$\Delta = \left\{ x = \sum_{i=0}^n \lambda_i a_i : \lambda \geq 0, \sum_i \lambda_i = 1 \right\} = \left\{ x = \sum_{i=1}^n \mu_i a_i : \mu \geq 0, \sum_{i=1}^n \mu_i \leq 1 \right\}.$$

We see that Δ is the image of the *standard full-dimensional simplex*

$$\left\{ \mu \in \mathbf{R}^n : \mu \geq 0, \sum_{i=1}^n \mu_i \leq 1 \right\}$$

under linear transformation $\mu \mapsto A\mu$, where A is the matrix with the columns a_1, \dots, a_n . The standard simplex clearly has a nonempty interior (comprised of all vectors $\mu > 0$ with $\sum_i \mu_i < 1$); since A is nonsingular (due to linear independence of a_1, \dots, a_n), multiplication by A maps open sets onto open ones, so that Δ has a nonempty interior. Since $\Delta \subset M$, the interior of M is nonempty. \square

(iii): We should prove that the closure of $\text{ri } M$ is exactly the same that the closure of M . In fact we shall prove even more:

Lemma B.1.1 *Let $x \in \text{ri } M$ and $y \in \text{cl } M$. Then all points from the half-segment $[x, y)$,*

$$[x, y) = \{z = (1 - \lambda)x + \lambda y : 0 \leq \lambda < 1\}$$

belong to the relative interior of M .

Proof of the Lemma. Let $\text{Aff}(M) = a + L$, L being linear subspace; then

$$M \subset \text{Aff}(M) = x + L.$$

Let B be the unit ball in L :

$$B = \{h \in L : \|h\|_2 \leq 1\}.$$

Since $x \in \text{ri } M$, there exists positive radius r such that

$$x + rB \subset M. \quad (\text{B.1.3})$$

Now let $\lambda \in [0, 1)$, and let $z = (1 - \lambda)x + \lambda y$. Since $y \in \text{cl } M$, we have $y = \lim_{i \rightarrow \infty} y_i$ for certain sequence of points from M . Setting $z_i = (1 - \lambda)x + \lambda y_i$, we get $z_i \rightarrow z$ as $i \rightarrow \infty$. Now, from (B.1.3) and the convexity of M it follows that the sets $Z_i = \{u = (1 - \lambda)x' + \lambda y_i : x' \in x + rB\}$ are contained in M ; clearly, Z_i is exactly the set $z_i + r'B$, where $r' = (1 - \lambda)r > 0$. Thus, z is the limit of sequence z_i , and r' -neighbourhood (in $\text{Aff}(M)$) of every one of the points z_i belongs to M . For every $r'' < r'$ and for all i such that z_i is close enough to z , the r' -neighbourhood of z_i contains the r'' -neighbourhood of z ; thus, a neighbourhood (in $\text{Aff}(M)$) of z belongs to M , whence $z \in \text{ri } M$. \blacksquare

A useful byproduct of Lemma B.1.1 is as follows:

Corollary B.1.2 *Let M be a convex set. Then every convex combination*

$$\sum_i \lambda_i x_i$$

of points $x_i \in \text{cl } M$ where at least one term with positive coefficient corresponds to $x_i \in \text{ri } M$ is in fact a point from $\text{ri } M$.

(iv): The statement is evidently true when M is empty, so assume that M is nonempty. The inclusion $\text{ri } M \subset \text{ri cl } M$ is evident, and all we need is to prove the inverse inclusion. Thus, let $z \in \text{ri cl } M$, and let us prove that $z \in \text{ri } M$. Let $x \in \text{ri } M$ (we already know that the latter set is nonempty). Consider the segment $[x, z]$; since z is in the relative interior of $\text{cl } M$, we can extend a little bit this segment through the point z , not leaving $\text{cl } M$, i.e., there exists $y \in \text{cl } M$ such that $z \in [x, y]$. We are done, since by Lemma B.1.1 from $z \in [x, y]$, with $x \in \text{ri } M$, $y \in \text{cl } M$, it follows that $z \in \text{ri } M$. ■

We see from the proof of Theorem B.1.1 that to get a closure of a (nonempty) convex set, it suffices to subject it to the “radial” closure, i.e., to take a point $x \in \text{ri } M$, take all rays in $\text{Aff}(M)$ starting at x and look at the intersection of such a ray l with M ; such an intersection will be a convex set on the line which contains a one-sided neighbourhood of x , i.e., is either a segment $[x, y_l]$, or the entire ray l , or a half-interval $[x, y_l)$. In the first two cases we should not do anything; in the third we should add y to M . After all rays are looked through and all “missed” endpoints y_l are added to M , we get the closure of M . To understand what is the role of convexity here, look at the *nonconvex* set of rational numbers from $[0, 1]$; the interior (\equiv relative interior) of this “highly percolated” set is empty, the closure is $[0, 1]$, and there is no way to restore the closure in terms of the interior.

B.2 Main theorems on convex sets

B.2.1 Caratheodory Theorem

Let us call the *affine dimension* (or simple *dimension*) of a nonempty set $M \subset \mathbf{R}^n$ (notation: $\dim M$) the affine dimension of $\text{Aff}(M)$.

Theorem B.2.1 [Caratheodory] *Let $M \subset \mathbf{R}^n$, and let $\dim \text{Conv}M = m$. Then every point $x \in \text{Conv}M$ is a convex combination of at most $m + 1$ points from M .*

Proof. Let $x \in \text{Conv}M$. By Proposition B.1.4 on the structure of convex hull, x is convex combination of certain points x_1, \dots, x_N from M :

$$x = \sum_{i=1}^N \lambda_i x_i, \quad [\lambda_i \geq 0, \sum_{i=1}^N \lambda_i = 1].$$

Let us choose among all these representations of x as a convex combination of points from M the one with the smallest possible N , and let it be the above combination. I claim that $N \leq m + 1$ (this claim leads to the desired statement). Indeed, if $N > m + 1$, then the system of $m + 1$ homogeneous equations

$$\begin{aligned} \sum_{i=1}^N \mu_i x_i &= 0 \\ \sum_{i=1}^N \mu_i &= 0 \end{aligned}$$

with N unknowns μ_1, \dots, μ_N has a nontrivial solution $\delta_1, \dots, \delta_N$:

$$\sum_{i=1}^N \delta_i x_i = 0, \quad \sum_{i=1}^N \delta_i = 0, \quad (\delta_1, \dots, \delta_N) \neq 0.$$

It follows that, for every real t ,

$$(*) \quad \sum_{i=1}^N [\lambda_i + t\delta_i]x_i = x.$$

What is to the left, is an affine combination of x_i 's. When $t = 0$, this is a convex combination - all coefficients are nonnegative. When t is large, this is *not* a convex combination, since some of δ_i 's are negative (indeed, not all of them are zero, and the sum of δ_i 's is 0). There exists, of course, the largest t for which the combination $(*)$ has nonnegative coefficients, namely

$$t^* = \min_{i:\delta_i < 0} \frac{\lambda_i}{|\delta_i|}.$$

For this value of t , the combination $(*)$ is with nonnegative coefficients, and at least one of the coefficients is zero; thus, we have represented x as a convex combination of less than N points from M , which contradicts the definition of N . ■

B.2.2 Radon Theorem

Theorem B.2.2 [Radon] *Let S be a set of at least $n + 2$ points x_1, \dots, x_N in \mathbf{R}^n . Then one can split the set into two nonempty subsets S_1 and S_2 with intersecting convex hulls: there exists partitioning $I \cup J = \{1, \dots, N\}$, $I \cap J = \emptyset$, of the index set $\{1, \dots, N\}$ into two nonempty sets I and J and convex combinations of the points $\{x_i, i \in I\}$, $\{x_j, j \in J\}$ which coincide with each other, i.e., there exist α_i , $i \in I$, and β_j , $j \in J$, such that*

$$\sum_{i \in I} \alpha_i x_i = \sum_{j \in J} \beta_j x_j; \quad \sum_i \alpha_i = \sum_j \beta_j = 1; \quad \alpha_i, \beta_j \geq 0.$$

Proof. Since $N > n + 1$, the homogeneous system of $n + 1$ scalar equations with N unknowns μ_1, \dots, μ_N

$$\begin{aligned} \sum_{i=1}^N \mu_i x_i &= 0 \\ \sum_{i=1}^N \mu_i &= 0 \end{aligned}$$

has a nontrivial solution $\lambda_1, \dots, \lambda_N$:

$$\sum_{i=1}^N \mu_i x_i = 0, \quad \sum_{i=1}^N \lambda_i = 0, \quad [(\lambda_1, \dots, \lambda_N) \neq 0].$$

Let $I = \{i : \lambda_i \geq 0\}$, $J = \{i : \lambda_i < 0\}$; then I and J are nonempty and form a partitioning of $\{1, \dots, N\}$. We have

$$a \equiv \sum_{i \in I} \lambda_i = \sum_{j \in J} (-\lambda_j) > 0$$

(since the sum of all λ 's is zero and not all λ 's are zero). Setting

$$\alpha_i = \frac{\lambda_i}{a}, \quad i \in I, \quad \beta_j = \frac{-\lambda_j}{a}, \quad j \in J,$$

we get

$$\alpha_i \geq 0, \quad \beta_j \geq 0, \quad \sum_{i \in I} \alpha_i = 1, \quad \sum_{j \in J} \beta_j = 1,$$

and

$$[\sum_{i \in I} \alpha_i x_i] - [\sum_{j \in J} \beta_j x_j] = a^{-1} \left([\sum_{i \in I} \lambda_i x_i] - [\sum_{j \in J} (-\lambda_j) x_j] \right) = a^{-1} \sum_{i=1}^N \lambda_i x_i = 0. \blacksquare$$

B.2.3 Helly Theorem

Theorem B.2.3 [Helly, I] *Let \mathcal{F} be a finite family of convex sets in \mathbf{R}^n . Assume that every $n + 1$ sets from the family have a point in common. Then all the sets have a point in common.*

Proof. Let us prove the statement by induction on the number N of sets in the family. The case of $N \leq n + 1$ is evident. Now assume that the statement holds true for all families with certain number $N \geq n + 1$ of sets, and let S_1, \dots, S_N, S_{N+1} be a family of $N + 1$ convex sets which satisfies the premise of the Helly Theorem; we should prove that the intersection of the sets S_1, \dots, S_N, S_{N+1} is nonempty.

Deleting from our $N + 1$ -set family the set S_i , we get N -set family which satisfies the premise of the Helly Theorem and thus, by the inductive hypothesis, the intersection of its members is nonempty:

$$(\forall i \leq N + 1) : T^i = S_1 \cap S_2 \cap \dots \cap S_{i-1} \cap S_{i+1} \cap \dots \cap S_{N+1} \neq \emptyset.$$

Let us choose a point x_i in the (nonempty) set T^i . We get $N + 1 \geq n + 2$ points from \mathbf{R}^n . By Radon's Theorem, we can partition the index set $\{1, \dots, N + 1\}$ into two nonempty subsets I and J in such a way that certain convex combination x of the points $x_i, i \in I$, is a convex combination of the points $x_j, j \in J$, as well. Let us verify that x belongs to all the sets S_1, \dots, S_{N+1} , which will complete the proof. Indeed, let i^* be an index from our index set; let us prove that $x \in S_{i^*}$. We have either $i^* \in I$, or $i^* \in J$. In the first case all the sets $T^j, j \in J$, are contained in S_{i^*} (since S_{i^*} participates in all intersections which give T^i with $i \neq i^*$). Consequently, all the points $x_j, j \in J$, belong to S_{i^*} , and therefore x , which is a convex combination of these points, also belongs to S_{i^*} (all our sets are convex!), as required. In the second case similar reasoning says that all the points $x_i, i \in I$, belong to S_{i^*} , and therefore x , which is a convex combination of these points, belongs to S_{i^*} . ■

Exercise B.9 *Let S_1, \dots, S_N be a family of N convex sets in \mathbf{R}^n , and let m be the affine dimension of $\text{Aff}(S_1 \cup \dots \cup S_N)$. Assume that every $m + 1$ sets from the family have a point in common. Prove that all sets from the family have a point in common.*

In the aforementioned version of the Helly Theorem we dealt with finite families of convex sets. To extend the statement to the case of infinite families, we need to strengthen slightly the assumption. The resulting statement is as follows:

Theorem B.2.4 [Helly, II] *Let \mathcal{F} be an arbitrary family of convex sets in \mathbf{R}^n . Assume that*

(a) *every $n + 1$ sets from the family have a point in common,*

and

(b) *every set in the family is closed, and the intersection of the sets from certain finite subfamily of the family is bounded (e.g., one of the sets in the family is bounded).*

Then all the sets from the family have a point in common.

Proof. By the previous theorem, all finite subfamilies of \mathcal{F} have nonempty intersections, and these intersections are convex (since intersection of a family of convex sets is convex, Theorem B.1.3); in view of (a) these intersections are also closed. Adding to \mathcal{F} all intersections of finite subfamilies of \mathcal{F} , we get a larger family \mathcal{F}' comprised of closed convex sets, and a finite subfamily of this larger family again has a nonempty intersection. Besides this, from (b) it follows that this new family contains a bounded set Q . Since all the sets are closed, the family of sets

$$\{Q \cap Q' : Q' \in \mathcal{F}\}$$

is a *nested family of compact sets* (i.e., a family of compact sets with nonempty intersection of sets from every finite subfamily); by the well-known Analysis theorem such a family has a nonempty intersection¹⁾. ■

B.2.4 Polyhedral representations and Fourier-Motzkin Elimination

B.2.4.1 B.2.4.A. Polyhedral representations

Recall that by definition a polyhedral set X in \mathbf{R}^n is the set of solutions of a finite system of nonstrict linear inequalities in variables $x \in \mathbf{R}^n$:

$$X = \{x \in \mathbf{R}^n : Ax \leq b\} = \{x \in \mathbf{R}^n : a_i^T x \leq b_i, 1 \leq i \leq m\}.$$

We shall call such a representation of X its *polyhedral description*. A polyhedral set always is convex and closed (Proposition B.1.1). Now let us introduce the notion of *polyhedral representation* of a set $X \subset \mathbf{R}^n$.

Definition B.2.1 We say that a set $X \subset \mathbf{R}^n$ is *polyhedrally representable*, if it admits a representation as follows:

$$X = \{x \in \mathbf{R}^n : \exists u \in \mathbf{R}^k : Ax + Bu \leq c\} \quad (\text{B.2.1})$$

where A, B are $m \times n$ and $m \times k$ matrices and $c \in \mathbf{R}^m$. A representation of X of the form of (B.2.1) is called a *polyhedral representation* of X , and variables u in such a representation are called *slack variables*.

Geometrically, a polyhedral representation of a set $X \subset \mathbf{R}^n$ is its representation as the projection $\{x : \exists u : (x, u) \in Y\}$ of a polyhedral set $Y = \{(x, u) : Ax + Bu \leq c\}$ in the space of $n + k$ variables $(x \in \mathbf{R}^n, u \in \mathbf{R}^k)$ under the linear mapping (the projection) $(x, u) \mapsto x : \mathbf{R}_{x,u}^{n+k} \rightarrow \mathbf{R}_x^n$ of the $n + k$ -dimensional space of (x, u) -variables (the space where Y lives) to the n -dimensional space of x -variables where X lives.

Note that every polyhedrally representable set is the image under linear mapping (even a projection) of a polyhedral, and thus convex, set. It follows that a *polyhedrally representable set definitely is convex* (Proposition B.1.6).

Examples: 1) Every polyhedral set $X = \{x \in \mathbf{R}^n : Ax \leq b\}$ is polyhedrally representable – a polyhedral description of X is nothing but a polyhedral representation with no slack variables

¹⁾here is the proof of this Analysis theorem: assume, on contrary, that the compact sets Q_α , $\alpha \in \mathcal{A}$, have empty intersection. Choose a set Q_{α^*} from the family; for every $x \in Q_{\alpha^*}$ there is a set Q^x in the family which does not contain x - otherwise x would be a common point of all our sets. Since Q^x is closed, there is an open ball V_x centered at x which does not intersect Q^x . The balls V_x , $x \in Q_{\alpha^*}$, form an open covering of the compact set Q_{α^*} , and therefore there exists a finite subcovering V_{x_1}, \dots, V_{x_N} of Q_{α^*} by the balls from the covering. Since Q^{x_i} does not intersect V_{x_i} , we conclude that the intersection of the finite subfamily $Q_{\alpha^*}, Q^{x_1}, \dots, Q^{x_N}$ is empty, which is a contradiction

($k = 0$). Vice versa, a polyhedral representation of a set X with no slack variables ($k = 0$) clearly is a polyhedral description of the set (which therefore is polyhedral).

2) Looking at the set $X = \{x \in \mathbf{R}^n : \sum_{i=1}^n |x_i| \leq 1\}$, we cannot say immediately whether it is or is not polyhedral; at least the initial description of X is *not* of the form $\{x : Ax \leq b\}$. However, X admits a polyhedral representation, e.g., the representation

$$X = \{x \in \mathbf{R}^n : \exists u \in \mathbf{R}^n : \underbrace{-u_i \leq x_i \leq u_i}_{\Leftrightarrow |x_i| \leq u_i}, 1 \leq i \leq n, \sum_{i=1}^n u_i \leq 1\}. \quad (\text{B.2.2})$$

Note that the set X in question can be described by a system of linear inequalities in x -variables only, namely, as

$$X = \{x \in \mathbf{R}^n : \sum_{i=1}^n \epsilon_i x_i \leq 1, \forall (\epsilon_i = \pm 1, 1 \leq i \leq n)\},$$

that is, X is polyhedral. However, the above polyhedral description of X (which in fact is minimal in terms of the number of inequalities involved) requires 2^n inequalities — an astronomically large number when n is just few tens. In contrast to this, the polyhedral representation (B.2.2) of the same set requires just n slack variables u and $2n + 1$ linear inequalities on x, u — the “complexity” of this representation is just linear in n .

3) Let a_1, \dots, a_m be given vectors in \mathbf{R}^n . Consider the conic hull of the finite set $\{a_1, \dots, a_m\}$ — the set $\text{Cone}\{a_1, \dots, a_m\} = \{x = \sum_{i=1}^m \lambda_i a_i : \lambda \geq 0\}$ (see Section B.1.4). It is absolutely unclear whether this set is polyhedral. In contrast to this, its polyhedral representation is immediate:

$$\begin{aligned} \text{Cone}\{a_1, \dots, a_m\} &= \{x \in \mathbf{R}^n : \exists \lambda \geq 0 : x = \sum_{i=1}^m \lambda_i a_i\} \\ &= \{x \in \mathbf{R}^n : \exists \lambda \in \mathbf{R}^m : \begin{cases} -\lambda \leq 0 \\ x - \sum_{i=1}^m \lambda_i a_i \leq 0 \\ -x + \sum_{i=1}^m \lambda_i a_i \leq 0 \end{cases}\} \end{aligned}$$

In other words, the original description of X is nothing but its polyhedral representation (in slight disguise), with λ_i 's in the role of slack variables.

B.2.4.2 B.2.4.B. Every polyhedrally representable set is polyhedral! (Fourier-Motzkin elimination)

A surprising and deep fact is that the situation in the Example 2) above is quite general:

Theorem B.2.5 *Every polyhedrally representable set is polyhedral.*

Proof: Fourier-Motskin Elimination. Recalling the definition of a polyhedrally representable set, our claim can be rephrased equivalently as follows: *the projection of a polyhedral set Y in a space $\mathbf{R}_{u,v}^{n+k}$ of (x, u) -variables on the subspace \mathbf{R}_x^n of x -variables is a polyhedral set in \mathbf{R}^n .* All we need is to prove this claim in the case of exactly one slack variable (since the projection which reduces the dimension by k — “kills k slack variables” — is the result of k subsequent projections, every one reducing the dimension by 1 (“killing one slack variable each”)).

Thus, let

$$Y = \{(x, u) \in \mathbf{R}^{n+1} : a_i^T x + b_i u \leq c_i, 1 \leq i \leq m\}$$

be a polyhedral set with n variables x and single variable u ; we want to prove that the projection

$$X = \{x : \exists u : Ax + bu \leq c\}$$

of Y on the space of x -variables is polyhedral. To see it, let us split the inequalities defining Y into three groups (some of them can be empty):

- “black” inequalities — those with $b_i = 0$; these inequalities do not involve u at all;
- “red” inequalities — those with $b_i > 0$. Such an inequality can be rewritten equivalently as $u \leq b_i^{-1}[c_i - a_i^T x]$, and it imposes a (depending on x) *upper bound* on u ;
- ”green” inequalities — those with $b_i < 0$. Such an inequality can be rewritten equivalently as $u \geq b_i^{-1}[c_i - a_i^T x]$, and it imposes a (depending on x) *lower bound* on u .

Now it is clear when $x \in X$, that is, when x can be extended, by some u , to a point (x, u) from Y : this is the case if and only if, first, x satisfies all black inequalities, and, second, the red upper bounds on u specified by x are compatible with the green lower bounds on u specified by x , meaning that every lower bound is \leq every upper bound (the latter is necessary and sufficient to be able to find a value of u which is \geq all lower bounds and \leq all upper bounds). Thus,

$$X = \left\{ x : \left\{ \begin{array}{l} a_i^T x \leq c_i \text{ for all “black” indexes } i - \text{those with } b_i = 0 \\ b_j^{-1}[c_j - a_j^T x] \leq b_k^{-1}[c_k - a_k^T x] \text{ for all “green” (i.e., with } b_j < 0) \text{ indexes } j \\ \text{and all “red” (i.e., with } b_k > 0) \text{ indexes } k \end{array} \right. \right\}.$$

We see that X is given by finitely many nonstrict linear inequalities in x -variables only, as claimed. ■

The outlined procedure for building polyhedral descriptions (i.e., polyhedral representations not involving slack variables) for projections of polyhedral sets is called *Fourier-Motzkin elimination*.

B.2.4.3 B.2.4.C. Some applications

As an immediate application of Fourier-Motzkin elimination, let us take a linear program $\min_x \{c^T x : Ax \leq b\}$ and look at the set T of values of the objective at all feasible solutions, if any:

$$T = \{t \in \mathbf{R} : \exists x : c^T x = t, Ax \leq b\}.$$

Rewriting the linear equality $c^T x = t$ as a pair of opposite inequalities, we see that T is polyhedrally representable, and the above definition of T is nothing but a polyhedral representation of this set, with x in the role of the vector of slack variables. By Fourier-Motzkin elimination, T is polyhedral — this set is given by a finite system of nonstrict linear inequalities in variable t only. As such, as it is immediately seen, T is

- either empty (meaning that the LP in question is infeasible),
- or is a below unbounded nonempty set of the form $\{t \in \mathbf{R} : -\infty \leq t \leq b\}$ with $b \in \mathbf{R} \cup \{+\infty\}$ (meaning that the LP is feasible and unbounded),
- or is a below bounded nonempty set of the form $\{t \in \mathbf{R} : -a \leq t \leq b\}$ with $a \in \mathbf{R}$ and $+\infty \geq b \geq a$. In this case, the LP is feasible and bounded, and a is its optimal value.

Note that given the list of linear inequalities defining T (this list can be built algorithmically by Fourier-Motzkin elimination as applied to the original polyhedral representation of T), we can easily detect which one of the above cases indeed takes place, i.e., to identify the feasibility and boundedness status of the LP and to find its optimal value. When it is finite (case 3 above),

we can use the Fourier-Motzkin elimination backward, starting with $t = a \in T$ and extending this value to a pair (t, x) with $t = a = c^T x$ and $Ax \leq b$, that is, we can augment the optimal value by an optimal solution. Thus, we can say that Fourier-Motzkin elimination is a finite Real Arithmetics algorithm which allows to check whether an LP is feasible and bounded, and when it is the case, allows to find the optimal value and an optimal solution. An unpleasant fact of life is that this algorithm is completely impractical, since the elimination process can blow up exponentially the number of inequalities. Indeed, from the description of the process it is clear that if a polyhedral set is given by m linear inequalities, then eliminating one variable, we can end up with as much as $m^2/4$ inequalities (this is what happens if there are $m/2$ red, $m/2$ green and no black inequalities). Eliminating the next variable, we again can “nearly square” the number of inequalities, and so on. Thus, the number of inequalities in the description of T can become astronomically large when even when the dimension of x is something like 10. The actual importance of Fourier-Motzkin elimination is of theoretical nature. For example, the LP-related reasoning we have just carried out shows that *every feasible and bounded LP program is solvable – has an optimal solution* (we shall revisit this result in more details in Section B.2.9.B). This is a fundamental fact for LP, and the above reasoning (even with the justification of the elimination “charged” to it) is the shortest and most transparent way to prove this fundamental fact. Another application of the fact that polyhedrally representable sets are polyhedral is the Homogeneous Farkas Lemma to be stated and proved in Section B.2.5.A; this lemma will be instrumental in numerous subsequent theoretical developments.

B.2.4.4 B.2.4.D. Calculus of polyhedral representations

The fact that polyhedral sets are exactly the same as polyhedrally representable ones does not nullify the notion of a polyhedral representation. The point is that a set can admit “quite compact” polyhedral representation involving slack variables and require astronomically large, completely meaningless for any practical purpose number of inequalities in its polyhedral description (think about the set (B.2.2) when $n = 100$). Moreover, polyhedral representations admit a kind of “fully algorithmic calculus.” Specifically, it turns out that all basic convexity-preserving operations (cf. Proposition B.1.6) as applied to polyhedral operands preserve polyhedrality; moreover, polyhedral representations of the results are readily given by polyhedral representations of the operands. Here is “algorithmic polyhedral analogy” of Proposition B.1.6:

1. Taking finite intersection: Let M_i , $1 \leq i \leq m$, be polyhedral sets in \mathbf{R}^n given by their polyhedral representations

$$M_i = \{x \in \mathbf{R}^n : \exists u^i \in \mathbf{R}^{k_i} : A_i x + B_i u^i \leq c^i\}, \quad 1 \leq i \leq m.$$

Then the intersection of the sets M_i is polyhedral with an explicit polyhedral representation, specifically,

$$\bigcap_{i=1}^m M_i = \{x \in \mathbf{R}^n : \exists u = (u^1, \dots, u^m) \in \mathbf{R}^{k_1+\dots+k_m} : \underbrace{A_i x + B_i u^i \leq c^i}_{\substack{\text{system of nonstrict linear} \\ \text{inequalities in } x, u}}, \quad 1 \leq i \leq m\}$$

2. Taking direct product: Let $M_i \subset \mathbf{R}^{n_i}$, $1 \leq i \leq m$, be polyhedral sets given by polyhedral representations

$$M_i = \{x^i \in \mathbf{R}^{n_i} : \exists u^i \in \mathbf{R}^{k_i} : A_i x^i + B_i u^i \leq c^i\}, \quad 1 \leq i \leq m.$$

Then the direct product $M_1 \times \dots \times M_m := \{x = (x^1, \dots, x^m) : x^i \in M_i, 1 \leq i \leq m\}$ of the sets is a polyhedral set with explicit polyhedral representation, specifically,

$$M_1 \times \dots \times M_m = \{x = (x^1, \dots, x^m) \in \mathbf{R}^{n_1+\dots+n_m} : \exists u = (u^1, \dots, u^m) \in \mathbf{R}^{k_1+\dots+k_m} : A_i x^i + B_i u^i \leq c^i, 1 \leq i \leq m\}.$$

3. Arithmetic summation and multiplication by reals: Let $M_i \subset \mathbf{R}^n, 1 \leq i \leq m$, be polyhedral sets given by polyhedral representations

$$M_i = \{x \in \mathbf{R}^n : \exists u^i \in \mathbf{R}^{k_i} : A_i x + B_i u^i \leq c^i\}, 1 \leq i \leq m,$$

and let $\lambda_1, \dots, \lambda_k$ be reals. Then the set $\lambda_1 M_1 + \dots + \lambda_m M_m := \{x = \lambda_1 x_1 + \dots + \lambda_m x_m : x_i \in M_i, 1 \leq i \leq m\}$ is polyhedral with explicit polyhedral representation, specifically,

$$\lambda_1 M_1 + \dots + \lambda_m M_m = \{x \in \mathbf{R}^n : \exists (x^i \in \mathbf{R}^n, u^i \in \mathbf{R}^{k_i}, 1 \leq i \leq m) : x \leq \sum_i \lambda_i x^i, x \geq \sum_i \lambda_i x^i, A_i x^i + B_i u^i \leq c^i, 1 \leq i \leq m\}.$$

4. Taking the image under an affine mapping: Let $M \subset \mathbf{R}^n$ be a polyhedral set given by polyhedral representation

$$M = \{x \in \mathbf{R}^n : \exists u \in \mathbf{R}^k : Ax + Bu \leq c\}$$

and let $\mathcal{P}(x) = Px + p : \mathbf{R}^n \rightarrow \mathbf{R}^m$ be an affine mapping. Then the image $\mathcal{P}(M) := \{y = Px + p : x \in M\}$ of M under the mapping is polyhedral set with explicit polyhedral representation, specifically,

$$\mathcal{P}(M) = \{y \in \mathbf{R}^m : \exists (x \in \mathbf{R}^n, u \in \mathbf{R}^k) : y \leq Px + p, y \geq Px + p, Ax + Bu \leq c\}.$$

5. Taking the inverse image under affine mapping: Let $M \subset \mathbf{R}^n$ be polyhedral set given by polyhedral representation

$$M = \{x \in \mathbf{R}^n : \exists u \in \mathbf{R}^k : Ax + Bu \leq c\}$$

and let $\mathcal{P}(y) = Py + p : \mathbf{R}^m \rightarrow \mathbf{R}^n$ be an affine mapping. Then the inverse image $\mathcal{P}^{-1}(M) := \{y : Py + p \in M\}$ of M under the mapping is polyhedral set with explicit polyhedral representation, specifically,

$$\mathcal{P}^{-1}(M) = \{y \in \mathbf{R}^m : \exists u : A(Py + p) + Bu \leq c\}.$$

Note that rules for intersection, taking direct products and taking inverse images, as applied to polyhedral descriptions of operands, lead to polyhedral descriptions of the results. In contrast to this, the rules for taking sums with coefficients and images under affine mappings heavily exploit the notion of polyhedral representation: even when the operands in these rules are given by polyhedral descriptions, there are no simple ways to point out polyhedral descriptions of the results.

Exercise B.10 Justify the above calculus rules.

Finally, we note that the problem of minimizing a linear form $c^T x$ over a set M given by polyhedral representation:

$$M = \{x \in \mathbf{R}^n : \exists u \in \mathbf{R}^k : Ax + Bu \leq c\}$$

can be immediately reduced to an explicit LP program, namely,

$$\min_{x,u} \left\{ c^T x : Ax + Bu \leq c \right\}.$$

A reader with some experience in Linear Programming definitely used a lot the above “calculus of polyhedral representations” when building LPs (perhaps without clear understanding of what in fact is going on, same as Molière’s Monsieur Jourdain all his life has been speaking prose without knowing it).

B.2.5 General Theorem on Alternative and Linear Programming Duality

Most of the contents of this Section is reproduced literally in the main body of the Notes (section 1.2). We repeat this text here to provide an option for smooth reading the Appendix.

B.2.5.1 B.2.5.A. Homogeneous Farkas Lemma

Let a_1, \dots, a_N be vectors from \mathbf{R}^n , and let a be another vector. Here we address the question: when a belongs to the cone spanned by the vectors a_1, \dots, a_N , i.e., when a can be represented as a linear combination of a_i with *nonnegative* coefficients? A *necessary* condition is evident: if

$$a = \sum_{i=1}^n \lambda_i a_i \quad [\lambda_i \geq 0, i = 1, \dots, N]$$

then every vector h which has nonnegative inner products with all a_i should also have nonnegative inner product with a :

$$a = \sum_i \lambda_i a_i \quad \& \quad \lambda_i \geq 0 \quad \forall i \quad \& \quad h^T a_i \geq 0 \quad \forall i \Rightarrow h^T a \geq 0.$$

The Homogeneous Farkas Lemma says that this evident necessary condition is also sufficient:

Lemma B.2.1 [Homogeneous Farkas Lemma] *Let a, a_1, \dots, a_N be vectors from \mathbf{R}^n . The vector a is a conic combination of the vectors a_i (linear combination with nonnegative coefficients) if and only if every vector h satisfying $h^T a_i \geq 0$, $i = 1, \dots, N$, satisfies also $h^T a \geq 0$. In other words, a homogeneous linear inequality*

$$a^T h \geq 0$$

in variable h is consequence of the system

$$a_i^T h \geq 0, \quad 1 \leq i \leq N$$

of homogeneous linear inequalities if and only if it can be obtained from the inequalities of the system by “admissible linear aggregation” – taking their weighted sum with nonnegative weights.

Proof. The necessity – the “only if” part of the statement – was proved before the Farkas Lemma was formulated. Let us prove the “if” part of the Lemma. Thus, assume that every vector h satisfying $h^T a_i \geq 0 \forall i$ satisfies also $h^T a \geq 0$, and let us prove that a is a conic combination of the vectors a_i .

An “intelligent” proof goes as follows. The set $\text{Cone}\{a_1, \dots, a_N\}$ of all conic combinations of a_1, \dots, a_N is polyhedrally representable (Example 3 in Section B.2.5.A.1) and as such is polyhedral (Theorem B.2.5):

$$\text{Cone}\{a_1, \dots, a_N\} = \{x \in \mathbf{R}^n : p_j^T x \geq b_j, 1 \leq j \leq J\}. \quad (!)$$

Observing that $0 \in \text{Cone}\{a_1, \dots, a_N\}$, we conclude that $b_j \leq 0$ for all j ; and since $\lambda a_i \in \text{Cone}\{a_1, \dots, a_N\}$ for every i and every $\lambda \geq 0$, we should have $\lambda p_j^T a_i \geq 0$ for all i, j and all $\lambda \geq 0$, whence $p_j^T a_i \geq 0$ for all i and j . For every j , relation $p_j^T a_i \geq 0$ for all i implies, by the premise of the statement we want to prove, that $p_j^T a \geq 0$, and since $b_j \leq 0$, we see that $p_j^T a \geq b_j$ for all j , meaning that a indeed belongs to $\text{Cone}\{a_1, \dots, a_N\}$ due to (!). ■

An interested reader can get a better understanding of the power of Fourier-Motzkin elimination, which ultimately is the basis for the above intelligent proof, by comparing this proof with the one based on Helley’s Theorem.

Proof based on Helley’s Theorem. As above, we assume that every vector h satisfying $h^T a_i \geq 0 \forall i$ satisfies also $h^T a \geq 0$, and we want to prove that a is a conic combination of the vectors a_i .

There is nothing to prove when $a = 0$ – the zero vector of course is a conic combination of the vectors a_i . Thus, from now on we assume that $a \neq 0$.

1⁰. Let

$$\Pi = \{h : a^T h = -1\},$$

and let

$$A_i = \{h \in \Pi : a_i^T h \geq 0\}.$$

Π is a hyperplane in \mathbf{R}^n , and every A_i is a polyhedral set contained in this hyperplane and is therefore convex.

2⁰. What we know is that the intersection of all the sets A_i , $i = 1, \dots, N$, is empty (since a vector h from the intersection would have nonnegative inner products with all a_i and the inner product -1 with a , and we are given that no such h exists). Let us choose the smallest, in the number of elements, of those sub-families of the family of sets A_1, \dots, A_N which still have empty intersection of their members; without loss of generality we may assume that this is the family A_1, \dots, A_k . Thus, the intersection of all k sets A_1, \dots, A_k is empty, but the intersection of every $k - 1$ sets from the family A_1, \dots, A_k is nonempty.

3⁰. We claim that

$$(A) \quad a \in \text{Lin}(\{a_1, \dots, a_k\});$$

$$(B) \quad \text{The vectors } a_1, \dots, a_k \text{ are linearly independent.}$$

(A) is easy: assuming that $a \notin E = \text{Lin}(\{a_1, \dots, a_k\})$, we conclude that the orthogonal projection f of the vector a onto the orthogonal complement E^\perp of E is nonzero. The inner product of f and a is the same as $f^T f$, i.e., is positive, while $f^T a_i = 0$, $i = 1, \dots, k$. Taking $h = -(f^T f)^{-1} f$, we see that $h^T a = -1$ and $h^T a_i = 0$, $i = 1, \dots, k$. In other words, h belongs to every set A_i , $i = 1, \dots, k$, by definition of these sets, and therefore the intersection of the sets A_1, \dots, A_k is nonempty, which is a contradiction.

(B) is given by the Helley Theorem I. (B) is evident when $k = 1$, since in this case linear dependence of a_1, \dots, a_k would mean that $a_1 = 0$; by (A), this implies that $a = 0$, which is not the case. Now let us prove (B) in the case of $k > 1$. Assume, on the contrary to what should be proven, that a_1, \dots, a_k are linearly dependent, so that the dimension of $E = \text{Lin}(\{a_1, \dots, a_k\})$ is certain $m < k$. We already know from A. that $a \in E$. Now let $A'_i = A_i \cap E$. We claim that every $k - 1$ of the sets A'_i have a nonempty intersection, while all k these sets have empty intersection. The second claim is evident – since the sets A_1, \dots, A_k have empty intersection, the same is the case with their parts A'_i . The first claim also is easily supported: let us take $k - 1$ of the dashed sets, say, A'_1, \dots, A'_{k-1} . By construction, the intersection of A_1, \dots, A_{k-1} is nonempty; let h be a vector from this intersection, i.e., a vector with nonnegative inner products with a_1, \dots, a_{k-1} and the product -1 with a . When replacing h with its orthogonal projection h' on E , we do not vary all these inner products, since these are products with vectors from E ; thus, h' also is a common point of A_1, \dots, A_{k-1} , and since this is a point from E , it is a common point of the dashed sets A'_1, \dots, A'_{k-1} as well.

Now we can complete the proof of (B): the sets A'_1, \dots, A'_k are convex sets belonging to the hyperplane $\Pi' = \Pi \cap E = \{h \in E : a^T h = -1\}$ (Π' indeed is a hyperplane in E , since $0 \neq a \in E$) in the m -dimensional linear subspace E . Π' is an affine subspace of the affine dimension $\ell = \dim E - 1 = m - 1 < k - 1$ (recall that we are in the situation when $m = \dim E < k$), and every $\ell + 1 \leq k - 1$ subsets from the family A'_1, \dots, A'_k have a nonempty intersection. From the Helley Theorem I (see Exercise B.9) it follows that all the sets A'_1, \dots, A'_k have a point in common, which, as we know, is not the case. The contradiction we have got proves that a_1, \dots, a_k are linearly independent.

4⁰. With (A) and (B) in our disposal, we can easily complete the proof of the “if” part of the Farkas Lemma. Specifically, by (A), we have

$$a = \sum_{i=1}^k \lambda_i a_i$$

with some real coefficients λ_i , and all we need is to prove that these coefficients are nonnegative. Assume, on the contrary, that, say, $\lambda_1 < 0$. Let us extend the (linearly independent in view of (B)) system of vectors a_1, \dots, a_k by vectors f_1, \dots, f_{n-k} to a basis in \mathbf{R}^n , and let $\xi_1(x)$ be the coordinates of a vector x in this basis. The function $\xi_1(x)$ is a linear form of x and therefore is the inner product with certain vector:

$$\xi_1(x) = f^T x \quad \forall x.$$

Now we have

$$f^T a = \xi_1(a) = \lambda_1 < 0$$

and

$$f^T a_i = \begin{cases} 1, & i = 1 \\ 0, & i = 2, \dots, k \end{cases}$$

so that $f^T a_i \geq 0$, $i = 1, \dots, k$. We conclude that a proper normalization of f – namely, the vector $|\lambda_1|^{-1} f$ – belongs to A_1, \dots, A_k , which is the desired contradiction – by construction, this intersection is empty. ■

B.2.5.2 B.2.5.B. General Theorem on Alternative

B.2.5.B.1 Certificates for solvability and insolvability. Consider a (finite) system of scalar inequalities with n unknowns. To be as general as possible, we do not assume for the time being the inequalities to be linear, and we allow for both non-strict and strict inequalities in the system, as well as for equalities. Since an equality can be represented by a pair of non-strict inequalities, our system can always be written as

$$f_i(x) \Omega_i 0, \quad i = 1, \dots, m, \tag{\mathcal{S}}$$

where every Ω_i is either the relation " $>$ " or the relation " \geq ".

The basic question about (\mathcal{S}) is

(?) Whether (\mathcal{S}) has a solution or not.

Knowing how to answer the question (?), we are able to answer many other questions. E.g., to verify whether a given real a is a lower bound on the optimal value c^* of (LP) is the same as to verify whether the system

$$\begin{cases} -c^T x + a > 0 \\ Ax - b \geq 0 \end{cases}$$

has no solutions.

The general question above is too difficult, and it makes sense to pass from it to a seemingly simpler one:

(??) How to certify that (\mathcal{S}) has, or does not have, a solution.

Imagine that you are very smart and know the correct answer to (?); how could you convince me that your answer is correct? What could be an “evident for everybody” validity certificate for your answer?

If your claim is that (\mathcal{S}) is solvable, a certificate could be just to point out a solution x^* to (\mathcal{S}) . Given this certificate, one can substitute x^* into the system and check whether x^* indeed is a solution.

Assume now that your claim is that (\mathcal{S}) has no solutions. What could be a “simple certificate” of this claim? How one could certify a negative statement? This is a highly nontrivial problem not just for mathematics; for example, in criminal law: how should someone accused in a murder prove his innocence? The “real life” answer to the question “how to certify a negative statement” is discouraging: such a statement normally *cannot* be certified (this is where the rule “a person is presumed innocent until proven guilty” comes from). In mathematics, however, the situation is different: in some cases there exist “simple certificates” of negative statements. E.g., in order to certify that (\mathcal{S}) has no solutions, it suffices to demonstrate that a consequence of (\mathcal{S}) is a contradictory inequality such as

$$-1 \geq 0.$$

For example, assume that $\lambda_i, i = 1, \dots, m$, are nonnegative weights. Combining inequalities from (\mathcal{S}) with these weights, we come to the inequality

$$\sum_{i=1}^m \lambda_i f_i(x) \Omega 0 \quad (\text{Comb}(\lambda))$$

where Ω is either " $>$ " (this is the case when the weight of at least one strict inequality from (\mathcal{S}) is positive), or " \geq " (otherwise). Since the resulting inequality, due to its origin, is a consequence of the system (\mathcal{S}) , i.e., it is satisfied by every solution to (\mathcal{S}) , it follows that if $(\text{Comb}(\lambda))$ has no solutions at all, we can be sure that (\mathcal{S}) has no solution. Whenever this is the case, we may treat the corresponding vector λ as a "simple certificate" of the fact that (\mathcal{S}) is infeasible.

Let us look what does the outlined approach mean when (\mathcal{S}) is comprised of *linear* inequalities:

$$(\mathcal{S}) : \quad \{a_i^T x \Omega_i b_i, i = 1, \dots, m\} \quad \left[\Omega_i = \begin{cases} " > " \\ " \geq " \end{cases} \right]$$

Here the "combined inequality" is linear as well:

$$(\text{Comb}(\lambda)) : \quad \left(\sum_{i=1}^m \lambda a_i \right)^T x \Omega \sum_{i=1}^m \lambda b_i$$

(Ω is " $>$ " whenever $\lambda_i > 0$ for at least one i with $\Omega_i = " > "$, and Ω is " \geq " otherwise). Now, when can a *linear* inequality

$$d^T x \Omega e$$

be contradictory? Of course, it can happen only when $d = 0$. Whether in this case the inequality is contradictory, it depends on what is the relation Ω : if $\Omega = " > "$, then the inequality is contradictory if and only if $e \geq 0$, and if $\Omega = " \geq "$, it is contradictory if and only if $e > 0$. We have established the following simple result:

Proposition B.2.1 *Consider a system of linear inequalities*

$$(\mathcal{S}) : \quad \begin{cases} a_i^T x > b_i, & i = 1, \dots, m_s, \\ a_i^T x \geq b_i, & i = m_s + 1, \dots, m. \end{cases}$$

with n -dimensional vector of unknowns x . Let us associate with (\mathcal{S}) two systems of linear inequalities and equations with m -dimensional vector of unknowns λ :

$$\mathcal{T}_I : \quad \begin{cases} (a) \quad \lambda \geq 0; \\ (b) \quad \sum_{i=1}^m \lambda_i a_i = 0; \\ (c_I) \quad \sum_{i=1}^m \lambda_i b_i \geq 0; \\ \hline (d_I) \quad \sum_{i=1}^{m_s} \lambda_i > 0. \end{cases}$$

$$\mathcal{T}_{II} : \quad \begin{cases} (a) \quad \lambda \geq 0; \\ (b) \quad \sum_{i=1}^m \lambda_i a_i = 0; \\ \hline (c_{II}) \quad \sum_{i=1}^m \lambda_i b_i > 0. \end{cases}$$

Assume that at least one of the systems \mathcal{T}_I , \mathcal{T}_{II} is solvable. Then the system (\mathcal{S}) is infeasible.

B.2.5.B.2 General Theorem on Alternative. Proposition B.2.1 says that in some cases it is easy to certify infeasibility of a linear system of inequalities: a “simple certificate” is a solution to another system of linear inequalities. Note, however, that the existence of a certificate of this latter type is to the moment only a *sufficient*, but not a *necessary*, condition for the infeasibility of (\mathcal{S}) . A fundamental result in the theory of linear inequalities is that the sufficient condition in question is in fact also necessary:

Theorem B.2.6 [General Theorem on Alternative] *In the notation from Proposition B.2.1, system (\mathcal{S}) has no solutions if and only if either \mathcal{T}_I , or \mathcal{T}_{II} , or both these systems, are solvable.*

Proof. GTA is a more or less straightforward corollary of the Homogeneous Farkas Lemma. Indeed, in view of Proposition B.2.1, all we need to prove is that if (\mathcal{S}) has no solution, then at least one of the systems \mathcal{T}_I , or \mathcal{T}_{II} is solvable. Thus, assume that (\mathcal{S}) has no solutions, and let us look at the consequences. Let us associate with (\mathcal{S}) the following system of homogeneous linear inequalities in variables x, τ, ϵ :

$$\begin{aligned} (a) \quad & \tau - \epsilon \geq 0 \\ (b) \quad & a_i^T x - b_i \tau - \epsilon \geq 0, i = 1, \dots, m_s \\ (c) \quad & a_i^T x - b_i \tau \geq 0, i = m_s + 1, \dots, m \end{aligned} \tag{B.2.3}$$

I claim that

(!) For every solution to (B.2.3), one has $\epsilon \leq 0$.

Indeed, assuming that (B.2.3) has a solution x, τ, ϵ with $\epsilon > 0$, we conclude from (B.2.3.a) that $\tau > 0$; from (B.2.3.b – c) it now follows that $\tau^{-1}x$ is a solution to (\mathcal{S}) , while the latter system is unsolvable.

Now, (!) says that the homogeneous linear inequality

$$-\epsilon \geq 0 \tag{B.2.4}$$

is a consequence of the system of homogeneous linear inequalities (B.2.3). By Homogeneous Farkas Lemma, it follows that there exist nonnegative weights $\nu, \lambda_i, i = 1, \dots, m$, such that the vector of coefficients of the variables x, τ, ϵ in the left hand side of (B.2.4) is linear combination, with the coefficients $\nu, \lambda_1, \dots, \lambda_m$, of the vectors of coefficients of the variables in the inequalities from (B.2.3):

$$\begin{aligned} (a) \quad & \sum_{i=1}^m \lambda_i a_i = 0 \\ (b) \quad & -\sum_{i=1}^m \lambda_i b_i + \nu = 0 \\ (c) \quad & -\sum_{i=1}^{m_s} \lambda_i - \nu = -1 \end{aligned} \tag{B.2.5}$$

Recall that by their origin, ν and all λ_i are nonnegative. Now, it may happen that $\lambda_1, \dots, \lambda_{m_s}$ are zero. In this case $\nu > 0$ by (B.2.5.c), and relations (B.2.5a – b) say that $\lambda_1, \dots, \lambda_m$ solve \mathcal{T}_{II} . In the remaining case (that is, when not all $\lambda_1, \dots, \lambda_{m_s}$ are zero, or, which is the same, when $\sum_{i=1}^{m_s} \lambda_i > 0$), the same relations say that $\lambda_1, \dots, \lambda_m$ solve \mathcal{T}_I . ■

B.2.5.B.3 Corollaries of the Theorem on Alternative. We formulate here explicitly two very useful principles following from the Theorem on Alternative:

A. A system of linear inequalities

$$a_i^T x \leq b_i, \quad i = 1, \dots, m$$

has no solutions if and only if one can combine the inequalities of the system in a linear fashion (i.e., multiplying the inequalities by nonnegative weights, adding the results and passing, if necessary, from an inequality $a^T x > b$ to the inequality $a^T x \geq b$) to get a contradictory inequality, namely, either the inequality $0^T x \geq 1$, or the inequality $0^T x > 0$.

B. A linear inequality

$$a_0^T x \leq b_0$$

is a consequence of a solvable system of linear inequalities

$$a_i^T x \leq b_i, \quad i = 1, \dots, m$$

if and only if it can be obtained by combining, in a linear fashion, the inequalities of the system and the trivial inequality $0 > -1$.

It should be stressed that the above principles are highly nontrivial and very deep. Consider, e.g., the following system of 4 linear inequalities with two variables u, v :

$$\begin{aligned} -1 &\leq u \leq 1 \\ -1 &\leq v \leq 1. \end{aligned}$$

From these inequalities it follows that

$$u^2 + v^2 \leq 2, \tag{!}$$

which in turn implies, by the Cauchy inequality, the linear inequality $u + v \leq 2$:

$$u + v = 1 \times u + 1 \times v \leq \sqrt{1^2 + 1^2} \sqrt{u^2 + v^2} \leq (\sqrt{2})^2 = 2. \tag{!!}$$

The concluding inequality is linear and is a consequence of the original system, but in the demonstration of this fact both steps (!) and (!!) are “highly nonlinear”. It is absolutely unclear a priori why the same consequence can, as it is stated by Principle **A**, be derived from the system in a linear manner as well [of course it can – it suffices just to add two inequalities $u \leq 1$ and $v \leq 1$].

Note that the Theorem on Alternative and its corollaries **A** and **B** heavily exploit the fact that we are speaking about *linear* inequalities. E.g., consider the following 2 quadratic and 2 linear inequalities with two variables:

- (a) $u^2 \geq 1;$
- (b) $v^2 \geq 1;$
- (c) $u \geq 0;$
- (d) $v \geq 0;$

along with the quadratic inequality

$$(e) \quad uv \geq 1.$$

The inequality (e) is clearly a consequence of (a) – (d). However, if we extend the system of inequalities (a) – (b) by all “trivial” (i.e., identically true) linear and quadratic inequalities with 2 variables, like $0 > -1$, $u^2 + v^2 \geq 0$, $u^2 + 2uv + v^2 \geq 0$, $u^2 - uv + v^2 \geq 0$, etc., and ask whether (e) can be derived in a *linear* fashion from the inequalities of the extended system, the answer will be negative. Thus, Principle **A** fails to be true already for quadratic inequalities (which is a great sorrow – otherwise there were no difficult problems at all!)

B.2.5.3 B.2.5.C. Application: Linear Programming Duality

We are about to use the Theorem on Alternative to obtain the basic results of the LP duality theory.

B.2.5.C.1 Dual to an LP program: the origin The motivation for constructing the problem dual to an LP program

$$c^* = \min_x \left\{ c^T x : Ax - b \geq 0 \right\} \quad \left[A = \begin{bmatrix} a_1^T \\ a_2^T \\ \dots \\ a_m^T \end{bmatrix} \in \mathbf{R}^{m \times n} \right] \quad (\text{LP})$$

is the desire to generate, in a systematic way, lower bounds on the optimal value c^* of (LP). An evident way to bound from below a given function $f(x)$ in the domain given by system of inequalities

$$g_i(x) \geq b_i, \quad i = 1, \dots, m, \quad (\text{B.2.6})$$

is offered by what is called the *Lagrange duality* and is as follows:

Lagrange Duality:

- Let us look at all inequalities which can be obtained from (B.2.6) by linear aggregation, i.e., at the inequalities of the form

$$\sum_i y_i g_i(x) \geq \sum_i y_i b_i \quad (\text{B.2.7})$$

with the “aggregation weights” $y_i \geq 0$. Note that the inequality (B.2.7), due to its origin, is valid on the entire set X of solutions of (B.2.6).

- Depending on the choice of aggregation weights, it may happen that the left hand side in (B.2.7) is $\leq f(x)$ for all $x \in \mathbf{R}^n$. Whenever it is the case, the right hand side $\sum_i y_i b_i$ of (B.2.7) is a lower bound on f in X .

Indeed, on X the quantity $\sum_i y_i b_i$ is a lower bound on $y_i g_i(x)$, and for y in question the latter function of x is everywhere $\leq f(x)$.

It follows that

- The optimal value in the problem

$$\max_y \left\{ \sum_i y_i b_i : \begin{array}{l} y \geq 0, \\ \sum_i y_i g_i(x) \leq f(x) \quad \forall x \in \mathbf{R}^n \end{array} \begin{array}{l} (a) \\ (b) \end{array} \right\} \quad (\text{B.2.8})$$

is a lower bound on the values of f on the set of solutions to the system (B.2.6).

Let us look what happens with the Lagrange duality when f and g_i are homogeneous linear functions: $f = c^T x$, $g_i(x) = a_i^T x$. In this case, the requirement (B.2.8.b) merely says that $c = \sum_i y_i a_i$ (or, which is the same, $A^T y = c$ due to the origin of A). Thus, problem (B.2.8) becomes the Linear Programming problem

$$\max_y \left\{ b^T y : A^T y = c, y \geq 0 \right\}, \quad (\text{LP}^*)$$

which is nothing but the LP dual of (LP).

By the construction of the dual problem,

[Weak Duality] *The optimal value in (LP^*) is less than or equal to the optimal value in (LP) .*

In fact, the “less than or equal to” in the latter statement is “equal”, provided that the optimal value c^* in (LP) is a number (i.e., (LP) is feasible and below bounded). To see that this indeed is the case, note that a real a is a lower bound on c^* if and only if $c^T x \geq a$ whenever $Ax \geq b$, or, which is the same, if and only if the system of linear inequalities

$$(\mathcal{S}_a) : \quad -c^T x > -a, Ax \geq b$$

has no solution. We know by the Theorem on Alternative that the latter fact means that some other system of linear equalities (more exactly, at least one of a certain pair of systems) does have a solution. More precisely,

(*) (\mathcal{S}_a) has no solutions if and only if at least one of the following two systems with $m + 1$ unknowns:

$$\mathcal{T}_I : \quad \left\{ \begin{array}{l} (a) \quad \lambda = (\lambda_0, \lambda_1, \dots, \lambda_m) \geq 0; \\ (b) \quad -\lambda_0 c + \sum_{i=1}^m \lambda_i a_i = 0; \\ \hline (c_I) \quad -\lambda_0 a + \sum_{i=1}^m \lambda_i b_i \geq 0; \\ (d_I) \quad \lambda_0 > 0, \end{array} \right.$$

or

$$\mathcal{T}_{II} : \quad \left\{ \begin{array}{l} (a) \quad \lambda = (\lambda_0, \lambda_1, \dots, \lambda_m) \geq 0; \\ (b) \quad -\lambda_0 c - \sum_{i=1}^m \lambda_i a_i = 0; \\ \hline (c_{II}) \quad -\lambda_0 a - \sum_{i=1}^m \lambda_i b_i > 0 \end{array} \right.$$

– has a solution.

Now assume that (LP) is feasible. We claim that *under this assumption (\mathcal{S}_a) has no solutions if and only if \mathcal{T}_I has a solution*.

The implication " \mathcal{T}_I has a solution $\Rightarrow (\mathcal{S}_a)$ has no solution" is readily given by the above remarks. To verify the inverse implication, assume that (\mathcal{S}_a) has no solutions and the system $Ax \leq b$ has a solution, and let us prove that then \mathcal{T}_I has a solution. If \mathcal{T}_I has no solution, then by (*) \mathcal{T}_{II} has a solution and, moreover, $\lambda_0 = 0$ for (every) solution to \mathcal{T}_{II} (since a solution to the latter system with $\lambda_0 > 0$ solves \mathcal{T}_I as well). But the fact that \mathcal{T}_{II} has a solution λ with $\lambda_0 = 0$ is independent of the values of a and c ; if this fact would take place, it would mean, by the same Theorem on Alternative, that, e.g., the following instance of (\mathcal{S}_a) :

$$0^T x \geq -1, Ax \geq b$$

has no solutions. The latter means that the system $Ax \geq b$ has no solutions – a contradiction with the assumption that (LP) is feasible. ■

Now, if \mathcal{T}_I has a solution, this system has a solution with $\lambda_0 = 1$ as well (to see this, pass from a solution λ to the one λ/λ_0 ; this construction is well-defined, since $\lambda_0 > 0$ for every solution to \mathcal{T}_I). Now, an $(m+1)$ -dimensional vector $\lambda = (1, y)$ is a solution to \mathcal{T}_I if and only if the m -dimensional vector y solves the system of linear inequalities and equations

$$\begin{aligned} y &\geq 0; \\ A^T y \equiv \sum_{i=1}^m y_i a_i &= c; \\ b^T y &\geq a \end{aligned} \tag{D}$$

Summarizing our observations, we come to the following result.

Proposition B.2.2 *Assume that system (D) associated with the LP program (LP) has a solution (y, a) . Then a is a lower bound on the optimal value in (LP). Vice versa, if (LP) is feasible and a is a lower bound on the optimal value of (LP), then a can be extended by a properly chosen m -dimensional vector y to a solution to (D).*

We see that the entity responsible for lower bounds on the optimal value of (LP) is the system (D): every solution to the latter system induces a bound of this type, and in the case when (LP) is feasible, all lower bounds can be obtained from solutions to (D). Now note that if (y, a) is a solution to (D), then the pair $(y, b^T y)$ also is a solution to the same system, and the lower bound $b^T y$ on c^* is not worse than the lower bound a . Thus, as far as lower bounds on c^* are concerned, we lose nothing by restricting ourselves to the solutions (y, a) of (D) with $a = b^T y$; the best lower bound on c^* given by (D) is therefore the optimal value of the problem $\max_y \{b^T y : A^T y = c, y \geq 0\}$, which is nothing but the dual to (LP) problem (LP*). Note that (LP*) is also a Linear Programming program.

All we know about the dual problem to the moment is the following:

Proposition B.2.3 *Whenever y is a feasible solution to (LP*), the corresponding value of the dual objective $b^T y$ is a lower bound on the optimal value c^* in (LP). If (LP) is feasible, then for every $a \leq c^*$ there exists a feasible solution y of (LP*) with $b^T y \geq a$.*

B.2.5.C.2 Linear Programming Duality Theorem. Proposition B.2.3 is in fact equivalent to the following

Theorem B.2.7 [Duality Theorem in Linear Programming] *Consider a linear programming program*

$$\min_x \{c^T x : Ax \geq b\} \tag{LP}$$

along with its dual

$$\max_y \{b^T y : A^T y = c, y \geq 0\} \tag{LP*}$$

Then

- 1) The duality is symmetric: the problem dual to dual is equivalent to the primal;
- 2) The value of the dual objective at every dual feasible solution is \leq the value of the primal objective at every primal feasible solution
- 3) The following 5 properties are equivalent to each other:

- (i) The primal is feasible and bounded below.
- (ii) The dual is feasible and bounded above.
- (iii) The primal is solvable.
- (iv) The dual is solvable.
- (v) Both primal and dual are feasible.

Whenever (i) \equiv (ii) \equiv (iii) \equiv (iv) \equiv (v) is the case, the optimal values of the primal and the dual problems are equal to each other.

Proof. 1) is quite straightforward: writing the dual problem (LP^*) in our standard form, we get

$$\min_y \left\{ -b^T y : \begin{bmatrix} I_m \\ A^T \\ -A^T \end{bmatrix} y - \begin{pmatrix} 0 \\ -c \\ c \end{pmatrix} \geq 0 \right\},$$

where I_m is the m -dimensional unit matrix. Applying the duality transformation to the latter problem, we come to the problem

$$\max_{\xi, \eta, \zeta} \left\{ 0^T \xi + c^T \eta + (-c)^T \zeta : \begin{array}{rcl} \xi & \geq & 0 \\ \eta & \geq & 0 \\ \zeta & \geq & 0 \\ \xi - A\eta + A\zeta & = & -b \end{array} \right\},$$

which is clearly equivalent to (LP) (set $x = \eta - \zeta$).

2) is readily given by Proposition B.2.3.

3):

(i) \Rightarrow (iv): If the primal is feasible and bounded below, its optimal value c^* (which of course is a lower bound on itself) can, by Proposition B.2.3, be (non-strictly) majorized by a quantity $b^T y^*$, where y^* is a feasible solution to (LP^*). In the situation in question, of course, $b^T y^* = c^*$ (by already proved item 2)); on the other hand, in view of the same Proposition B.2.3, the optimal value in the dual is $\leq c^*$. We conclude that the optimal value in the dual is attained and is equal to the optimal value in the primal.

(iv) \Rightarrow (ii): evident;

(ii) \Rightarrow (iii): This implication, in view of the primal-dual symmetry, follows from the implication (i) \Rightarrow (iv).

(iii) \Rightarrow (i): evident.

We have seen that (i) \equiv (ii) \equiv (iii) \equiv (iv) and that the first (and consequently each) of these 4 equivalent properties implies that the optimal value in the primal problem is equal to the optimal value in the dual one. All which remains is to prove the equivalence between (i)–(iv), on one hand, and (v), on the other hand. This is immediate: (i)–(iv), of course, imply (v); vice versa, in the case of (v) the primal is not only feasible, but also bounded below (this is an immediate consequence of the feasibility of the dual problem, see 2)), and (i) follows. ■

An immediate corollary of the LP Duality Theorem is the following necessary and sufficient optimality condition in LP:

Theorem B.2.8 [Necessary and sufficient optimality conditions in linear programming] *Consider an LP program (LP) along with its dual (LP*). A pair (x, y) of primal and dual feasible solutions is comprised of optimal solutions to the respective problems if and only if*

$$y_i[Ax - b]_i = 0, \quad i = 1, \dots, m, \quad [\text{complementary slackness}]$$

likewise as if and only if

$$c^T x - b^T y = 0 \quad [\text{zero duality gap}]$$

Indeed, the “zero duality gap” optimality condition is an immediate consequence of the fact that the value of primal objective at every primal feasible solution is \geq the value of the dual objective at every dual feasible solution, while the optimal values in the primal and the dual are equal to each other, see Theorem B.2.7. The equivalence between the “zero duality gap” and the “complementary slackness” optimality conditions is given by the following computation: whenever x is primal feasible and y is dual feasible, the products $y_i[Ax - b]_i$, $i = 1, \dots, m$, are nonnegative, while the sum of these products is precisely the duality gap:

$$y^T [Ax - b] = (A^T y)^T x - b^T y = c^T x - b^T y.$$

Thus, the duality gap can vanish at a primal-dual feasible pair (x, y) if and only if all products $y_i[Ax - b]_i$ for this pair are zeros.

B.2.6 Separation Theorem

B.2.6.1 B.2.6.A. Separation: definition

Recall that a *hyperplane* M in \mathbf{R}^n is, by definition, an affine subspace of the dimension $n - 1$. By Proposition A.3.7, hyperplanes are exactly the same as level sets of nontrivial linear forms:

$$\begin{aligned} M \subset \mathbf{R}^n \text{ is a hyperplane} \\ \Updownarrow \\ \exists a \in \mathbf{R}^n, b \in \mathbf{R}, a \neq 0 : \quad M = \{x \in \mathbf{R}^n : a^T x = b\} \end{aligned}$$

We can, consequently, associate with the hyperplane (or, better to say, with the associated linear form a ; this form is defined uniquely, up to multiplication by a nonzero real) the following sets:

- “upper” and “lower” open half-spaces $M^{++} = \{x \in \mathbf{R}^n : a^T x > b\}$, $M^{--} = \{x \in \mathbf{R}^n : a^T x < b\}$;

these sets clearly are convex, and since a linear form is continuous, and the sets are given by strict inequalities on the value of a continuous function, they indeed are open.

Note that since a is uniquely defined by M , up to multiplication by a nonzero real, these open half-spaces are uniquely defined by the hyperplane, up to swapping the “upper” and the “lower” ones (which half-space is “upper”, it depends on the particular choice of a);

- “upper” and “lower” closed half-spaces $M^+ = \{x \in \mathbf{R}^n : a^T x \geq b\}$, $M^- = \{x \in \mathbf{R}^n : a^T x \leq b\}$;

these are also convex sets, now closed (since they are given by non-strict inequalities on the value of a continuous function). It is easily seen that the closed upper/lower half-space is the closure of the corresponding open half-space, and M itself is the boundary (i.e., the complement of the interior to the closure) of all four half-spaces.

It is clear that our half-spaces and M itself partition \mathbf{R}^n :

$$\mathbf{R}^n = M^{--} \cup M \cup M^{++}$$

(partitioning by disjoint sets),

$$\mathbf{R}^n = M^- \cup M^+$$

(M is the intersection of the right hand side sets).

Now we define the basic notion of separation of two convex sets T and S by a hyperplane.

Definition B.2.2 [separation] *Let S, T be two nonempty convex sets in \mathbf{R}^n .*

- *A hyperplane*

$$M = \{x \in \mathbf{R}^n : a^T x = b\} \quad [a \neq 0]$$

is said to separate S and T , if, first,

$$S \subset \{x : a^T x \leq b\}, \quad T \subset \{x : a^T x \geq b\}$$

(i.e., S and T belong to the opposite closed half-spaces into which M splits \mathbf{R}^n), and, second, at least one of the sets S, T is not contained in M itself:

$$S \cup T \not\subset M.$$

The separation is called strong, if there exist $b', b'', b' < b < b''$, such that

$$S \subset \{x : a^T x \leq b'\}, \quad T \subset \{x : a^T x \geq b''\}.$$

- *A linear form $a \neq 0$ is said to separate (strongly separate) S and T , if for properly chosen b the hyperplane $\{x : a^T x = b\}$ separates (strongly separates) S and T .*
- *We say that S and T can be (strongly) separated, if there exists a hyperplane which (strongly) separates S and T .*

E.g.,

- the hyperplane $\{x : a^T x \equiv x_2 - x_1 = 1\}$ in \mathbf{R}^2 strongly separates convex polyhedral sets $T = \{x \in \mathbf{R}^2 : 0 \leq x_1 \leq 1, 3 \leq x_2 \leq 5\}$ and $S = \{x \in \mathbf{R}^2 : x_2 = 0; x_1 \geq -1\}$;
- the hyperplane $\{x : a^T x \equiv x = 1\}$ in \mathbf{R}^1 separates (but not strongly separates) the convex sets $S = \{x \leq 1\}$ and $T = \{x \geq 1\}$;
- the hyperplane $\{x : a^T x \equiv x_1 = 0\}$ in \mathbf{R}^2 separates (but not strongly separates) the sets $S = \{x \in \mathbf{R}^2 : x_1 < 0, x_2 \geq -1/x_1\}$ and $T = \{x \in \mathbf{R}^2 : x_1 > 0, x_2 > 1/x_1\}$;
- the hyperplane $\{x : a^T x \equiv x_2 - x_1 = 1\}$ in \mathbf{R}^2 does not separate the convex sets $S = \{x \in \mathbf{R}^2 : x_2 \geq 1\}$ and $T = \{x \in \mathbf{R}^2 : x_2 = 0\}$;
- the hyperplane $\{x : a^T x \equiv x_2 = 0\}$ in \mathbf{R}^2 does not separate the sets $S = \{x \in \mathbf{R}^2 : x_2 = 0, x_1 \leq -1\}$ and $T = \{x \in \mathbf{R}^2 : x_2 = 0, x_1 \geq 1\}$.

The following Exercise presents an equivalent description of separation:

Exercise B.11 Let S, T be nonempty convex sets in \mathbf{R}^n . Prove that a linear form a separates S and T if and only if

$$\sup_{x \in S} a^T x \leq \inf_{y \in T} a^T y$$

and

$$\inf_{x \in S} a^T x < \sup_{y \in T} a^T y.$$

This separation is strong if and only if

$$\sup_{x \in S} a^T x < \inf_{y \in T} a^T y.$$

Exercise B.12 Whether the sets $S = \{x \in \mathbf{R}^2 : x_1 > 0, x_2 \geq 1/x_1\}$ and $T = \{x \in \mathbf{R}^2 : x_1 < 0, x_2 \geq -1/x_1\}$ can be separated? Whether they can be strongly separated?

B.2.6.2 B.2.6.B. Separation Theorem

Theorem B.2.9 [Separation Theorem] Let S and T be nonempty convex sets in \mathbf{R}^n .

(i) S and T can be separated if and only if their relative interiors do not intersect: $\text{ri } S \cap \text{ri } T = \emptyset$.

(ii) S and T can be strongly separated if and only if the sets are at a positive distance from each other:

$$\text{dist}(S, T) \equiv \inf\{\|x - y\|_2 : x \in S, y \in T\} > 0.$$

In particular, if S, T are closed nonempty non-intersecting convex sets and one of these sets is compact, S and T can be strongly separated.

Proof takes several steps.

(i), Necessity. Assume that S, T can be separated, so that for certain $a \neq 0$ we have

$$\inf_{x \in S} a^T x \leq \inf_{y \in T} a^T y; \quad \inf_{x \in S} a^T x < \sup_{y \in T} a^T y. \quad (\text{B.2.9})$$

We should lead to a contradiction the assumption that $\text{ri } S$ and $\text{ri } T$ have in common certain point \bar{x} . Assume that it is the case; then from the first inequality in (B.2.9) it is clear that \bar{x} maximizes the linear function $f(x) = a^T x$ on S and simultaneously minimizes this function on T . Now, we have the following simple and important

Lemma B.2.2 A linear function $f(x) = a^T x$ can attain its maximum/minimum over a convex set Q at a point $x \in \text{ri } Q$ if and only if the function is constant on Q .

Proof. "if" part is evident. To prove the "only if" part, let $\bar{x} \in \text{ri } Q$ be, say, a minimizer of f over Q and y be an arbitrary point of Q ; we should prove that $f(\bar{x}) = f(y)$. There is nothing to prove if $y = \bar{x}$, so let us assume that $y \neq \bar{x}$. Since $\bar{x} \in \text{ri } Q$, the segment $[y, \bar{x}]$, which is contained in M , can be extended a little bit through the point \bar{x} , not leaving M (since $\bar{x} \in \text{ri } Q$), so that there exists $z \in Q$ such that $\bar{x} \in [y, z]$, i.e., $\bar{x} = (1 - \lambda)y + \lambda z$ with certain $\lambda \in (0, 1]$; since $y \neq \bar{x}$, we have in fact $\lambda \in (0, 1)$. Since f is linear, we have

$$f(\bar{x}) = (1 - \lambda)f(y) + \lambda f(z);$$

since $f(\bar{x}) \leq \min\{f(y), f(z)\}$ and $0 < \lambda < 1$, this relation can be satisfied only when $f(\bar{x}) = f(y) = f(z)$. \square

By Lemma B.2.2, $f(x) = f(\bar{x})$ on S and on T , so that $f(\cdot)$ is constant on $S \cup T$, which yields the desired contradiction with the second inequality in (B.2.9). \square

(i), Sufficiency. The proof of sufficiency part of the Separation Theorem is much more instructive. There are several ways to prove it, and I choose the one which goes via the Homogeneous Farkas Lemma B.2.1, which is extremely important in its own right.

(i), Sufficiency, Step 1: Separation of a convex polytope and a point outside the polytope. Let us start with seemingly very particular case of the Separation Theorem – the one where S is the convex hull of points x_1, \dots, x_N , and T is a singleton $T = \{x\}$ which does not belong to S . We intend to prove that in this case there exists a linear form which separates x and S ; in fact we shall prove even the existence of strong separation.

Let us associate with n -dimensional vectors x_1, \dots, x_N, x the $(n+1)$ -dimensional vectors $a = \begin{pmatrix} x \\ 1 \end{pmatrix}$ and $a_i = \begin{pmatrix} x_i \\ 1 \end{pmatrix}$, $i = 1, \dots, N$. I claim that a does not belong to the conic hull of a_1, \dots, a_N . Indeed, if a would be representable as a linear combination of a_1, \dots, a_N with nonnegative coefficients, then, looking at the last, $(n+1)$ -st, coordinates in such a representation, we would conclude that the sum of coefficients should be 1, so that the representation, actually, represents x as a convex combination of x_1, \dots, x_N , which was assumed to be impossible.

Since a does not belong to the conic hull of a_1, \dots, a_N , by the Homogeneous Farkas Lemma (Lemma B.2.1) there exists a vector $h = \begin{pmatrix} f \\ \alpha \end{pmatrix} \in \mathbf{R}^{n+1}$ which “separates” a and a_1, \dots, a_N in the sense that

$$h^T a > 0, \quad h^T a_i \leq 0, \quad i = 1, \dots, N,$$

whence, of course,

$$h^T a > \max_i h^T a_i.$$

Since the components in all the inner products $h^T a, h^T a_i$ coming from the $(n+1)$ -st coordinates are equal to each other, we conclude that the n -dimensional component f of h separates x and x_1, \dots, x_N :

$$f^T x > \max_i f^T x_i.$$

Since for every convex combination $y = \sum_i \lambda_i x_i$ of the points x_i one clearly has $f^T y \leq \max_i f^T x_i$, we conclude, finally, that

$$f^T x > \max_{y \in \text{Conv}(\{x_1, \dots, x_N\})} f^T y,$$

so that f strongly separates $T = \{x\}$ and $S = \text{Conv}(\{x_1, \dots, x_N\})$. \square

(i), Sufficiency, Step 2: Separation of a convex set and a point outside of the set. Now consider the case when S is an arbitrary nonempty convex set and $T = \{x\}$ is a singleton outside S (the difference with Step 1 is that now S is not assumed to be a polytope).

First of all, without loss of generality we may assume that S contains 0 (if it is not the case, we may subject S and T to translation $S \mapsto p + S$, $T \mapsto p + T$ with $p \in -S$). Let L be the linear span of S . If $x \notin L$, the separation is easy: taking as f the orthogonal to L component of x , we shall get

$$f^T x = f^T f > 0 = \max_{y \in S} f^T y,$$

so that f strongly separates S and $T = \{x\}$.

It remains to consider the case when $x \in L$. Since $S \subset L$, $x \in L$ and $x \notin S$, L is a nonzero linear subspace; w.l.o.g., we can assume that $L = \mathbf{R}^n$.

Let $\Sigma = \{h : \|h\|_2 = 1\}$ be the unit sphere in $L = \mathbf{R}^n$. This is a closed and bounded set in \mathbf{R}^n (boundedness is evident, and closedness follows from the fact that $\|\cdot\|_2$ is continuous). Consequently, Σ is a compact set. Let us prove that there exists $f \in \Sigma$ which separates x and S in the sense that

$$f^T x \geq \sup_{y \in S} f^T y. \quad (\text{B.2.10})$$

Assume, on the contrary, that no such f exists, and let us lead this assumption to a contradiction. Under our assumption for every $h \in \Sigma$ there exists $y_h \in S$ such that

$$h^T y_h > h^T x.$$

Since the inequality is strict, it immediately follows that there exists a neighbourhood U_h of the vector h such that

$$(h')^T y_h > (h')^T x \quad \forall h' \in U_h. \quad (\text{B.2.11})$$

The family of open sets $\{U_h\}_{h \in \Sigma}$ covers Σ ; since Σ is compact, we can find a finite subfamily U_{h_1}, \dots, U_{h_N} of the family which still covers Σ . Let us take the corresponding points $y_1 = y_{h_1}, y_2 = y_{h_2}, \dots, y_N = y_{h_N}$ and the polytope $S' = \text{Conv}(\{y_1, \dots, y_N\})$ spanned by the points. Due to the origin of y_i , all of them are points from S ; since S is convex, the polytope S' is contained in S and, consequently, does not contain x . By Step 1, x can be strongly separated from S' : there exists a such that

$$a^T x > \sup_{y \in S'} a^T y. \quad (\text{B.2.12})$$

By normalization, we may also assume that $\|a\|_2 = 1$, so that $a \in \Sigma$. Now we get a contradiction: since $a \in \Sigma$ and U_{h_1}, \dots, U_{h_N} form a covering of Σ , a belongs to certain U_{h_i} . By construction of U_{h_i} (see (B.2.11)), we have

$$a^T y_i \equiv a^T y_{h_i} > a^T x,$$

which contradicts (B.2.12) – recall that $y_i \in S'$.

The contradiction we get proves that there exists $f \in \Sigma$ satisfying (B.2.10). We claim that f separates S and $\{x\}$; in view of (B.2.10), all we need to verify our claim is to show that the linear form $f(y) = f^T y$ is non-constant on $S \cup T$, which is evident: we are in the situation when $0 \in S$ and $L \equiv \text{Lin}(S) = \mathbf{R}^n$ and $f \neq 0$, so that $f(y)$ is non-constant already on S . \square

Mathematically oriented reader should take into account that the simple-looking reasoning underlying Step 2 in fact brings us into a completely new world. Indeed, the considerations at Step 1 and in the proof of Homogeneous Farkas Lemma are “pure arithmetic” – we never used things like convergence, compactness, etc., and used rational arithmetic only – no square roots, etc. It means that the Homogeneous Farkas Lemma and the result stated at Step 1 remain valid if we, e.g., replace our universe \mathbf{R}^n with the space \mathbf{Q}^n of n -dimensional rational vectors (those with rational coordinates; of course, the multiplication by reals in this space should be restricted to multiplication by rationals). The “rational” Farkas Lemma or the possibility to separate a rational vector from a “rational” polytope by a rational linear form, which is the “rational” version of the result of Step 1, definitely are of interest (e.g., for Integer Programming). In contrast to these “purely arithmetic” considerations, at Step 2 we used compactness – something heavily exploiting the fact that our universe is \mathbf{R}^n and not, say, \mathbf{Q}^n (in the latter space bounded and closed sets not necessarily are compact). Note also that we could not avoid things like compactness arguments at Step 2, since the very fact we are proving is true in \mathbf{R}^n but not in \mathbf{Q}^n . Indeed, consider the “rational plane” – the

universe comprised of all 2-dimensional vectors with rational entries, and let S be the half-plane in this rational plane given by the linear inequality

$$x_1 + \alpha x_2 \leq 0,$$

where α is irrational. S clearly is a “convex set” in \mathbf{Q}^2 ; it is immediately seen that a point outside this set cannot be separated from S by a rational linear form.

(i), Sufficiency, Step 3: Separation of two nonempty and non-intersecting convex sets. Now we are ready to prove that two nonempty and non-intersecting convex sets S and T can be separated. To this end consider the arithmetic difference

$$\Delta = S - T = \{x - y : x \in S, y \in T\}.$$

By Proposition B.1.6.3, Δ is convex (and, of course, nonempty) set; since S and T do not intersect, Δ does not contain 0. By Step 2, we can separate Δ and $\{0\}$: there exists $f \neq 0$ such that

$$f^T 0 = 0 \geq \sup_{z \in \Delta} f^T z \quad \& \quad f^T 0 > \inf_{z \in \Delta} f^T z.$$

In other words,

$$0 \geq \sup_{x \in S, y \in T} [f^T x - f^T y] \quad \& \quad 0 > \inf_{x \in S, y \in T} [f^T x - f^T y],$$

which clearly means that f separates S and T . ■

(i), Sufficiency, Step 4: Separation of nonempty convex sets with non-intersecting relative interiors. Now we are able to complete the proof of the “if” part of the Separation Theorem. Let S and T be two nonempty convex sets with non-intersecting relative interiors; we should prove that S and T can be properly separated. This is immediate: as we know from Theorem B.1.1, the sets $S' = \text{ri } S$ and $T' = \text{ri } T$ are nonempty and convex; since we are given that they do not intersect, they can be separated by Step 3: there exists f such that

$$\inf_{x \in T'} f^T x \geq \sup_{y \in S'} f^T y \quad \& \quad \sup_{x \in T'} f^T x > \inf_{y \in S'} f^T y. \quad (\text{B.2.13})$$

It is immediately seen that in fact f separates S and T . Indeed, the quantities in the left and the right hand sides of the first inequality in (B.2.13) clearly remain unchanged when we replace S' with $\text{cl } S'$ and T' with $\text{cl } T'$; by Theorem B.1.1, $\text{cl } S' = \text{cl } S \supset S$ and $\text{cl } T' = \text{cl } T \supset T$, and we get $\inf_{x \in T} f^T x = \inf_{x \in T'} f^T x$, and similarly $\sup_{y \in S} f^T y = \sup_{y \in S'} f^T y$. Thus, we get from (B.2.13)

$$\inf_{x \in T} f^T x \geq \sup_{y \in S} f^T y.$$

It remains to note that $T' \subset T$, $S' \subset S$, so that the second inequality in (B.2.13) implies that

$$\sup_{x \in T} f^T x > \inf_{y \in S} f^T y. \quad \square$$

(ii), Necessity: prove yourself.

(ii), **Sufficiency:** Assuming that $\rho \equiv \inf\{\|x - y\|_2 : x \in S, y \in T\} > 0$, consider the sets $S' = \{x : \inf_{y \in S} \|x - y\|_2 \leq \rho\}$. Note that S' is convex along with S (Example B.1.4) and that $S' \cap T = \emptyset$ (why?). By (i), S' and T can be separated, and if f is a linear form which separates S' and T , then the same form strongly separates S and T (why?). The “in particular” part of (ii) readily follows from the just proved statement due to the fact that if two closed nonempty sets in \mathbf{R}^n do not intersect and one of them is compact, then the sets are at positive distance from each other (why?). ■

Exercise B.13 Derive the statement in Remark B.1.1 from the Separation Theorem.

Exercise B.14 Implement the following alternative approach to the proof of Separation Theorem:

1. Prove that if x is a point in \mathbf{R}^n and S is a nonempty closed convex set in \mathbf{R}^n , then the problem

$$\min_y \{\|x - y\|_2 : y \in S\}$$

has a unique optimal solution \bar{x} .

2. In the situation of 1), prove that if $x \notin S$, then the linear form $e = x - \bar{x}$ strongly separates $\{x\}$ and S :

$$\max_{y \in S} e^T y = e^T \bar{x} = e^T x - e^T e < e^T x,$$

thus getting a direct proof of the possibility to separate strongly a nonempty closed convex set and a point outside this set.

3. Derive from 2) the Separation Theorem.

B.2.6.3 B.2.6.C. Supporting hyperplanes

By the Separation Theorem, a closed and nonempty convex set M is the intersection of all closed half-spaces containing M . Among these half-spaces, the most interesting are the “extreme” ones – those with the boundary hyperplane touching M . The notion makes sense for an arbitrary (not necessarily closed) convex set, but we shall use it for closed sets only, and include the requirement of closedness in the definition:

Definition B.2.3 [Supporting plane] Let M be a convex closed set in \mathbf{R}^n , and let x be a point from the relative boundary of M . A hyperplane

$$\Pi = \{y : a^T y = a^T x\} \quad [a \neq 0]$$

is called supporting to M at x , if it separates M and $\{x\}$, i.e., if

$$a^T x \geq \sup_{y \in M} a^T y \quad \& \quad a^T x > \inf_{y \in M} a^T y. \quad (\text{B.2.14})$$

Note that since x is a point from the relative boundary of M and therefore belongs to $\text{cl } M = M$, the first inequality in (B.2.14) in fact is equality. Thus, an equivalent definition of a supporting plane is as follows:

Let M be a closed convex set and x be a relative boundary point of M . The hyperplane $\{y : a^T y = a^T x\}$ is called supporting to M at x , if the linear form $a(y) = a^T y$ attains its maximum on M at the point x and is nonconstant on M .

E.g., the hyperplane $\{x_1 = 1\}$ in \mathbf{R}^n clearly is supporting to the unit Euclidean ball $\{x : |x| \leq 1\}$ at the point $x = e_1 = (1, 0, \dots, 0)$.

The most important property of a supporting plane is its existence:

Proposition B.2.4 [Existence of supporting hyperplane] *Let M be a convex closed set in \mathbf{R}^n and x be a point from the relative boundary of M . Then*

(i) *There exists at least one hyperplane which is supporting to M at x ;*

(ii) *If Π is supporting to M at x , then the intersection $M \cap \Pi$ is of affine dimension less than the one of M (recall that the affine dimension of a set is, by definition, the affine dimension of the affine hull of the set).*

Proof. (i) is easy: if x is a point from the relative boundary of M , then it is outside the relative interior of M and therefore $\{x\}$ and $\text{ri } M$ can be separated by the Separation Theorem; the separating hyperplane is exactly the desired supporting to M at x hyperplane.

To prove (ii), note that if $\Pi = \{y : a^T y = a^T x\}$ is supporting to M at $x \in \partial_{\text{ri}} M$, then the set $M' = M \cap \Pi$ is a nonempty (it contains x) convex set, and the linear form $a^T y$ is constant on M' and therefore (why?) on $\text{Aff}(M')$. At the same time, the form is nonconstant on M by definition of a supporting plane. Thus, $\text{Aff}(M')$ is a proper (less than the entire $\text{Aff}(M)$) subset of $\text{Aff}(M)$, and therefore the affine dimension of $\text{Aff}(M')$ (i.e., the affine dimension of M') is less than the affine dimension of $\text{Aff}(M)$ (i.e., than the affine dimension of M). ²⁾ ■

B.2.7 Polar of a convex set and Milutin-Dubovitski Lemma

B.2.7.1 B.2.7.A. Polar of a convex set

Let M be a nonempty convex set in \mathbf{R}^n . The polar $\text{Polar}(M)$ of M is the set of all linear forms which do not exceed 1 on M , i.e., the set of all vectors a such that $a^T x \leq 1$ for all $x \in M$:

$$\text{Polar}(M) = \{a : a^T x \leq 1 \forall x \in M\}.$$

For example, $\text{Polar}(\mathbf{R}^n) = \{0\}$, $\text{Polar}(\{0\}) = \mathbf{R}^n$; if L is a liner subspace in \mathbf{R}^n , then $\text{Polar}(L) = L^\perp$ (why?).

The following properties of the polar are evident:

1. $0 \in \text{Polar}(M)$;
2. $\text{Polar}(M)$ is convex;
3. $\text{Polar}(M)$ is closed.

It turns out that these properties characterize polars:

Proposition B.2.5 *Every closed convex set M containing the origin is polar, specifically, it is polar of its polar:*

$$\begin{gathered} M \text{ is closed and convex, } 0 \in M \\ \Updownarrow \\ M = \text{Polar}(\text{Polar}(M)) \end{gathered}$$

²⁾ In the latter reasoning we used the following fact: if $P \subset Q$ are two affine subspaces, then the affine dimension of P is \leq the one of Q , with \leq being = if and only if $P = Q$. Please prove this fact

Proof. All we need is to prove that if M is closed and convex and $0 \in M$, then $M = \text{Polar}(\text{Polar}(M))$. By definition,

$$y \in \text{Polar}(M), x \in M \Rightarrow y^T x \leq 1,$$

so that $M \subset \text{Polar}(\text{Polar}(M))$. To prove that this inclusion is in fact equality, assume, on the contrary, that there exists $\bar{x} \in \text{Polar}(\text{Polar}(M)) \setminus M$. Since M is nonempty, convex and closed and $\bar{x} \notin M$, the point \bar{x} can be strongly separated from M (Separation Theorem, (ii)). Thus, for appropriate b one has

$$b^T \bar{x} > \sup_{x \in M} b^T x.$$

Since $0 \in M$, the left hand side quantity in this inequality is positive; passing from b to a proportional vector $a = \lambda b$ with appropriately chosen positive λ , we may ensure that

$$a^T \bar{x} > 1 \geq \sup_{x \in M} a^T x.$$

This is the desired contradiction, since the relation $1 \geq \sup_{x \in M} a^T x$ implies that $a \in \text{Polar}(M)$, so that the relation $a^T \bar{x} > 1$ contradicts the assumption that $\bar{x} \in \text{Polar}(\text{Polar}(M))$. ■

Exercise B.15 Let M be a convex set containing the origin, and let M' be the polar of M . Prove the following facts:

1. $\text{Polar}(M) = \text{Polar}(\text{cl } M)$;
2. M is bounded if and only if $0 \in \text{int } M'$;
3. $\text{int } M \neq \emptyset$ if and only if M' does not contain straight lines;
4. M is a closed cone if and only if M' is a closed cone. If M is a cone (not necessarily closed), then

$$M' = \{a : a^T x \leq 0 \forall x \in M\}. \quad (\text{B.2.15})$$

B.2.7.2 B.2.7.B. Dual cone

Let $M \subset \mathbf{R}^n$ be a cone. By Exercise B.15.4, the polar M' of M is a closed cone given by (B.2.15). The set $M_* = -M'$ (which also is a closed cone), that is, the set

$$M_* = \{a : a^T x \geq 0 \forall x \in M\}$$

of all vectors which have nonnegative inner products with all vectors from M , is called the cone dual to M . By Proposition B.2.5 and Exercise B.15.4, the family of closed cones in \mathbf{R}^n is closed with respect to passing to a dual cone, and the duality is symmetric: for a closed cone M , M_* also is a closed cone, and $(M_*)_* = M$.

Exercise B.16 Let M be a closed cone in \mathbf{R}^n , and M_* be its dual cone. Prove that

1. M is pointed (i.e., does not contain lines) if and only if M_* has a nonempty interior. Derive from this fact that M is a closed pointed cone with a nonempty interior if and only if the dual cone has the same properties.
2. Prove that $a \in \text{int } M_*$ if and only if $a^T x > 0$ for all nonzero vectors $x \in M$.

B.2.7.3 B.2.7.C. Dubovitski-Milutin Lemma

Let M_1, \dots, M_k be cones (not necessarily closed), and M be their intersection; of course, M also is a cone. How to compute the cone dual to M ?

Proposition B.2.6 *Let M_1, \dots, M_k be cones. The cone M' dual to the intersection M of the cones M_1, \dots, M_k contains the arithmetic sum \widetilde{M} of the cones M'_1, \dots, M'_k dual to M_1, \dots, M_k . If all the cones M_1, \dots, M_k are closed, then M' is equal to $\text{cl } \widetilde{M}$. In particular, for closed cones M_1, \dots, M_k , M' coincides with \widetilde{M} if and only if the latter set is closed.*

Proof. Whenever $a_i \in M'_i$ and $x \in M$, we have $a_i^T x \geq 0$, $i = 1, \dots, k$, whence $(a_1 + \dots + a_k)^T x \geq 0$. Since the latter relation is valid for all $x \in M$, we conclude that $a_1 + \dots + a_k \in M'$. Thus, $M \subset M'$.

Now assume that the cones M_1, \dots, M_k are closed, and let us prove that $M = \text{cl } \widetilde{M}$. Since M' is closed and we have seen that $\widetilde{M} \subset M'$, all we should prove is that if $a \in M'$, then $a \in \widetilde{M} = \text{cl } \widetilde{M}$ as well. Assume, on the contrary, that $a \in M' \setminus \widetilde{M}$. Since the set \widetilde{M} clearly is a cone, its closure \widetilde{M} is a closed cone; by assumption, a does not belong to this closed cone and therefore, by Separation Theorem (ii), a can be strongly separated from \widetilde{M} and therefore – from $\widetilde{M} \subset \widehat{M}$. Thus, for some x one has

$$a^T x < \inf_{b \in \widetilde{M}} b^T x = \inf_{a_i \in M'_i, i=1, \dots, k} (a_1 + \dots + a_k)^T x = \sum_{i=1}^k \inf_{a_i \in M'_i} a_i^T x. \quad (\text{B.2.16})$$

From the resulting inequality it follows that $\inf_{a_i \in M'_i} a_i^T x > -\infty$; since M'_i is a cone, the latter is possible if and only if $\inf_{a_i \in M'_i} a_i^T x = 0$, i.e., if and only if for every i one has $x \in \text{Polar}(M'_i) = M_i$ (recall that the cones M_i are closed). Thus, $x \in M_i$ for all i , and the concluding quantity in (B.2.16) is 0. We see that $x \in M = \cap_i M_i$, and that (B.2.16) reduces to $a^T x < 0$. This contradicts the inclusion $a \in M'$. ■

Note that in general \widetilde{M} can be non-closed even when all the cones M_1, \dots, M_k are closed. Indeed, take $k = 2$, and let M'_1 be the ice-cream cone $\{(x, y, z) \in \mathbf{R}^3 : z \geq \sqrt{x^2 + y^2}\}$, and M'_2 be the ray $\{z = x \leq 0, y = 0\}$ in bR^3 . Observe that the points from $\widetilde{M} \equiv M'_1 + M'_2$ are exactly the points of the form $(x - t, y, z - t)$ with $t \geq 0$ and $z \geq \sqrt{x^2 + y^2}$. In particular, for x positive the points $(0, 1, \sqrt{x^2 + 1} - x) = (x - x, 1, \sqrt{x^2 + 1} - x)$ belong to \widetilde{M} ; as $x \rightarrow \infty$, these points converge to $p = (0, 1, 0)$, and thus $p \in \text{cl } \widetilde{M}$. On the other hand, there clearly do not exist x, y, z, t with $t \geq 0$ and $z \geq \sqrt{x^2 + y^2}$ such that $(x - t, y, z - t) = (0, 1, 0)$, that is, $p \notin \widetilde{M}$.

Dubovitski-Milutin Lemma presents a simple sufficient condition for \widetilde{M} to be closed and thus to coincide with M' :

Proposition B.2.7 [Dubovitski-Milutin Lemma] *Let M_1, \dots, M_k be cones such that M_k is closed and the set $M_k \cap \text{int } M_1 \cap \text{int } M_2 \cap \dots \cap \text{int } M_{k-1}$ is nonempty, and let $M = M_1 \cap \dots \cap M_k$. Let also M'_i be the cones dual to M_i . Then*

$$(i) \text{ cl } M = \bigcap_{i=1}^k \text{cl } M_i;$$

(ii) the cone $\widetilde{M} = M'_1 + \dots + M'_k$ is closed, and thus coincides with the cone M' dual to $\text{cl } M$ (or, which is the same by Exercise B.15.1, with the cone dual to M). In other words, every linear form which is nonnegative on M can be represented as a sum of k linear forms which are nonnegative on the respective cones M_1, \dots, M_k .

Proof. (i): We should prove that under the premise of the Dubovitski-Milutin Lemma, $\text{cl } M = \bigcap_i \text{cl } M_i$. The right hand side here contains M and is closed, so that all we should prove is that

every point x in $\bigcap_{i=1}^k \text{cl } M_i$ is the limit of an appropriate sequence $x_t \in M$. By premise of the Lemma, there exists a point $\bar{x} \in M_k \cap \text{int } M_1 \cap \text{int } M_2 \cap \dots \cap \text{int } M_{k-1}$; setting $x_t = t^{-1}\bar{x} + (1-t^{-1})x$, $t = 1, 2, \dots$, we get a sequence converging to x as $t \rightarrow \infty$; at the same time, $x_t \in M_k$ (since x, \bar{x} are in $\text{cl } M_k = M_k$) and $x_t \in M_i$ for every $i < k$ (by Lemma B.1.1; note that for $i < k$ one has $\bar{x} \in \text{int } M_i$ and $x \in \text{cl } M_i$), and thus $x_t \in M$. Thus, every point $x \in \bigcap_{i=1}^k \text{cl } M_i$ is the limit of a sequence from M . \square

(ii): Under the premise of the Lemma, when replacing the cones M_1, \dots, M_k with their closures, we do not vary the polars M'_i of the cones (and thus do not vary \tilde{M}) and replace the intersection of the sets M_1, \dots, M_k with its closure (by (i)), thus not varying the polar of the intersection. And of course when replacing the cones M_1, \dots, M_k with their closures, we preserve the premise of Lemma. Thus, we lose nothing when assuming, in addition to the premise of Lemma, that the cones M_1, \dots, M_k are closed. To prove the lemma for closed cones M_1, \dots, M_k , we use induction in $k \geq 2$.

Base $k = 2$: Let a sequence $\{f_t + g_t\}_{t=1}^\infty$ with $f_t \in M'_1$ and $g_t \in M'_2$ converge to certain h ; we should prove that $h = f + g$ for appropriate $f \in M'_1$ and $g \in M'_2$. To achieve our goal, it suffices to verify that for an appropriate subsequence t_j of indices there exists $f \equiv \lim_{j \rightarrow \infty} f_{t_j}$.

Indeed, if this is the case, then $g = \lim_{j \rightarrow \infty} g_{t_j}$ also exists (since $f_t + g_t \rightarrow h$ as $t \rightarrow \infty$ and $f + g = h$; besides this, $f \in M'_1$ and $g \in M'_2$, since both the cones in question are closed). In order to verify the existence of the desired subsequence, it suffices to lead to a contradiction the assumption that $\|f_t\|_2 \rightarrow \infty$ as $t \rightarrow \infty$. Let the latter assumption be true. Passing to a subsequence, we may assume that the unit vectors $\phi_t = f_t/\|f_t\|_2$ have a limit ϕ as $t \rightarrow \infty$; since M'_1 is a closed cone, ϕ is a unit vector from M'_1 . Now, since $f_t + g_t \rightarrow h$ as $t \rightarrow \infty$, we have $\phi = \lim_{t \rightarrow \infty} f_t/\|f_t\|_2 = -\lim_{t \rightarrow \infty} g_t/\|f_t\|_2$ (recall that $\|f_t\|_2 \rightarrow \infty$ as $t \rightarrow \infty$, whence $h/\|f_t\|_2 \rightarrow 0$ as $t \rightarrow \infty$). We see that the vector $-\phi$ belongs to M'_2 . Now, by assumption M_2 intersects the interior of the cone M_1 ; let \bar{x} be a point in this intersection. We have $\phi^T \bar{x} \geq 0$ (since $\bar{x} \in M_1$ and $\phi \in M'_1$) and $\phi^T \bar{x} \leq 0$ (since $-\phi \in M'_2$ and $\bar{x} \in M_2$). We conclude that $\phi^T \bar{x} = 0$, which contradicts the facts that $0 \neq \phi \in M'_1$ and $\bar{x} \in \text{int } M_1$ (see Exercise B.16.2). \square

Inductive step: Assume that the statement we are proving is valid in the case of $k-1 \geq 2$ cones, and let M_1, \dots, M_k be k cones satisfying the premise of the Dubovitski-Milutin Lemma. By this premise, the cone $M^1 = M_1 \cap \dots \cap M_{k-1}$ has a nonempty interior, and M_k intersects this interior. Applying to the pair of cones M^1, M_k the already proved 2-cone version of the Lemma, we see that the set $(M^1)' + M'_k$ is closed; here $(M^1)'$ is the cone dual to M^1 . Further, the cones M_1, \dots, M_{k-1} satisfy the premise of the $(k-1)$ -cone version of the Lemma; by inductive hypothesis, the set $M'_1 + \dots + M'_{k-1}$ is closed and therefore, by Proposition B.2.6, equals to $(M^1)'$. Thus, $M'_1 + \dots + M'_k = (M^1)' + M'_k$, and we have seen that the latter set is closed. ■

B.2.8 Extreme points and Krein-Milman Theorem

Supporting planes are useful tool to prove existence of extreme points of convex sets. Geometrically, an extreme point of a convex set M is a point in M which cannot be obtained as a convex

combination of other points of the set; and the importance of the notion comes from the fact (which we shall prove in the mean time) that the set of all extreme points of a “good enough” convex set M is the “shortest worker’s instruction for building the set” – this is the smallest set of points for which M is the convex hull.

B.2.8.1 B.2.8.A. Extreme points: definition

The exact definition of an extreme point is as follows:

Definition B.2.4 [extreme points] *Let M be a nonempty convex set in \mathbf{R}^n . A point $x \in M$ is called an extreme point of M , if there is no nontrivial (of positive length) segment $[u, v] \in M$ for which x is an interior point, i.e., if the relation*

$$x = \lambda u + (1 - \lambda)v$$

with $\lambda \in (0, 1)$ and $u, v \in M$ is valid if and only if

$$u = v = x.$$

E.g., the extreme points of a segment are exactly its endpoints; the extreme points of a triangle are its vertices; the extreme points of a (closed) circle on the 2-dimensional plane are the points of the circumference.

An equivalent definitions of an extreme point is as follows:

Exercise B.17 *Let M be a convex set and let $x \in M$. Prove that*

1. *x is extreme if and only if the only vector h such that $x \pm h \in M$ is the zero vector;*
2. *x is extreme if and only if the set $M \setminus \{x\}$ is convex.*

B.2.8.2 B.2.8.B. Krein-Milman Theorem

It is clear that a convex set M not necessarily possesses extreme points; as an example you may take the open unit ball in \mathbf{R}^n . This example is not interesting – the set in question is not closed; when replacing it with its closure, we get a set (the closed unit ball) with plenty of extreme points – these are all points of the boundary. There are, however, *closed* convex sets which do not possess extreme points – e.g., a line or an affine subspace of larger dimension. A nice fact is that the absence of extreme points in a closed convex set M always has the standard reason – the set contains a line. Thus, a closed and nonempty convex set M which does not contain lines for sure possesses extreme points. And if M is nonempty convex compact set, it possesses a quite representative set of extreme points – their convex hull is the entire M .

Theorem B.2.10 *Let M be a closed and nonempty convex set in \mathbf{R}^n . Then*

- (i) *The set $\text{Ext}(M)$ of extreme points of M is nonempty if and only if M does not contain lines;*
- (ii) *If M is bounded, then M is the convex hull of its extreme points:*

$$M = \text{Conv}(\text{Ext}(M)),$$

so that every point of M is a convex combination of the points of $\text{Ext}(M)$.

Part (ii) of this theorem is the finite-dimensional version of the famous *Krein-Milman Theorem*.

Proof. Let us start with (i). The "only if" part is easy, due to the following simple

Lemma B.2.3 *Let M be a closed convex set in \mathbf{R}^n . Assume that for some $\bar{x} \in M$ and $h \in \mathbf{R}^n$ M contains the ray*

$$\{\bar{x} + th : t \geq 0\}$$

starting at \bar{x} with the direction h . Then M contains also all parallel rays starting at the points of M :

$$(\forall x \in M) : \{x + th : t \geq 0\} \subset M.$$

In particular, if M contains certain line, then it contains also all parallel lines passing through the points of M .

Comment. For a closed convex set M , the set of all directions h such that $x + th \in M$ for some x and all $t \geq 0$ (i.e., by Lemma – such that $x + th \in M$ for all $x \in M$ and all $t \geq 0$) is called *the recessive cone of M* [notation: $\text{Rec}(M)$]. With Lemma B.2.3 it is immediately seen (prove it!) that $\text{Rec}(M)$ indeed is a closed cone, and that

$$M + \text{Rec}(M) = M.$$

Directions from $\text{Rec}(M)$ are called recessive for M .

Proof of the lemma is immediate: if $x \in M$ and $\bar{x} + th \in M$ for all $t \geq 0$, then, due to convexity, for any fixed $\tau \geq 0$ we have

$$\epsilon(\bar{x} + \frac{\tau}{\epsilon}h) + (1 - \epsilon)x \in M$$

for all $\epsilon \in (0, 1)$. As $\epsilon \rightarrow +0$, the left hand side tends to $x + \tau h$, and since M is closed, $x + \tau h \in M$ for every $\tau \geq 0$. \square

Exercise B.18 *Let M be a closed nonempty convex set. Prove that $\text{Rec}(M) \neq \{0\}$ if and only if M is unbounded.*

Lemma B.2.3, of course, resolves all our problems with the "only if" part. Indeed, here we should prove that if M possesses extreme points, then M does not contain lines, or, which is the same, that if M contains lines, then it has no extreme points. But the latter statement is immediate: if M contains a line, then, by Lemma, there is a line in M passing through every given point of M , so that no point can be extreme. \square

Now let us prove the "if" part of (i). Thus, from now on we assume that M does not contain lines; our goal is to prove that then M possesses extreme points. Let us start with the following

Lemma B.2.4 *Let Q be a nonempty closed convex set, \bar{x} be a relative boundary point of Q and Π be a hyperplane supporting to Q at \bar{x} . Then all extreme points of the nonempty closed convex set $\Pi \cap Q$ are extreme points of Q .*

Proof of the Lemma. First, the set $\Pi \cap Q$ is closed and convex (as an intersection of two sets with these properties); it is nonempty, since it contains \bar{x} (Π contains \bar{x}

due to the definition of a supporting plane, and Q contains \bar{x} due to the closedness of Q). Second, let a be the linear form associated with Π :

$$\Pi = \{y : a^T y = a^T \bar{x}\},$$

so that

$$\inf_{x \in Q} a^T x < \sup_{x \in Q} a^T x = a^T \bar{x} \quad (\text{B.2.17})$$

(see Proposition B.2.4). Assume that y is an extreme point of $\Pi \cap Q$; what we should do is to prove that y is an extreme point of Q , or, which is the same, to prove that

$$y = \lambda u + (1 - \lambda)v$$

for some $u, v \in Q$ and $\lambda \in (0, 1)$ is possible only if $y = u = v$. To this end it suffices to demonstrate that under the above assumptions $u, v \in \Pi \cap Q$ (or, which is the same, to prove that $u, v \in \Pi$, since the points are known to belong to Q); indeed, we know that y is an extreme point of $\Pi \cap Q$, so that the relation $y = \lambda u + (1 - \lambda)v$ with $\lambda \in (0, 1)$ and $u, v \in \Pi \cap Q$ does imply $y = u = v$.

To prove that $u, v \in \Pi$, note that since $y \in \Pi$ we have

$$a^T y = a^T \bar{x} \geq \max\{a^T u, a^T v\}$$

(the concluding inequality follows from (B.2.17)). On the other hand,

$$a^T y = \lambda a^T u + (1 - \lambda) a^T v;$$

combining these observations and taking into account that $\lambda \in (0, 1)$, we conclude that

$$a^T y = a^T u = a^T v.$$

But these equalities imply that $u, v \in \Pi$. \square

Equipped with the Lemma, we can easily prove (i) by induction on the dimension of the convex set M (recall that this is nothing but the affine dimension of the affine span of M , i.e., the linear dimension of the linear subspace L such that $\text{Aff}(M) = a + L$).

There is nothing to do if the dimension of M is zero, i.e., if M is a point – then, of course, $M = \text{Ext}(M)$. Now assume that we already have proved the nonemptiness of $\text{Ext}(T)$ for all nonempty closed and not containing lines convex sets T of certain dimension k , and let us prove that the same statement is valid for the sets of dimension $k + 1$. Let M be a closed convex nonempty and not containing lines set of dimension $k + 1$. Since M does not contain lines and is of positive dimension, it differs from $\text{Aff}(M)$ and therefore it possesses a relative boundary point \bar{x} ³⁾. According to Proposition B.2.4, there exists a hyperplane $\Pi = \{x : a^T x = a^T \bar{x}\}$ which supports M at \bar{x} :

$$\inf_{x \in M} a^T x < \max_{x \in M} a^T x = a^T \bar{x}.$$

³⁾Indeed, there exists $z \in \text{Aff}(M) \setminus M$, so that the points

$$x_\lambda = x + \lambda(z - x)$$

(x is an arbitrary fixed point of M) do not belong to M for some $\lambda \geq 1$, while $x_0 = x$ belongs to M . The set of those $\lambda \geq 0$ for which $x_\lambda \in M$ is therefore nonempty and bounded from above; this set clearly is closed (since M is closed). Thus, there exists the largest $\lambda = \lambda^*$ for which $x_\lambda \in M$. We claim that x_{λ^*} is a relative boundary point of M . Indeed, by construction this is a point from M . If it would be a point from the relative interior of M , then all the points x_λ with close to λ^* and greater than λ^* values of λ would also belong to M , which contradicts the origin of λ^* .

By the same Proposition, the set $T = \Pi \cap M$ (which is closed, convex and nonempty) is of affine dimension less than the one of M , i.e., of the dimension $\leq k$. T clearly does not contain lines (since even the larger set M does not contain lines). By the inductive hypothesis, T possesses extreme points, and by Lemma B.2.4 all these points are extreme also for M . The inductive step is completed, and (i) is proved. \square

Now let us prove (ii). Thus, let M be nonempty, convex, closed and bounded; we should prove that

$$M = \text{Conv}(\text{Ext}(M)).$$

What is immediately seen is that the right hand side set is contained in the left hand side one. Thus, all we need is to prove that every $x \in M$ is a convex combination of points from $\text{Ext}(M)$. Here we again use induction on the affine dimension of M . The case of 0-dimensional set M (i.e., a point) is trivial. Assume that the statement in question is valid for all k -dimensional convex closed and bounded sets, and let M be a convex closed and bounded set of dimension $k+1$. Let $x \in M$; to represent x as a convex combination of points from $\text{Ext}(M)$, let us pass through x an arbitrary line $\ell = \{x + \lambda h : \lambda \in \mathbf{R}\}$ ($h \neq 0$) in the affine span $\text{Aff}(M)$ of M . Moving along this line from x in each of the two possible directions, we eventually leave M (since M is bounded); as it was explained in the proof of (i), it means that there exist nonnegative λ_+ and λ_- such that the points

$$\bar{x}_\pm = x + \lambda_\pm h$$

both belong to the relative boundary of M . Let us verify that \bar{x}_\pm are convex combinations of the extreme points of M (this will complete the proof, since x clearly is a convex combination of the two points \bar{x}_\pm). Indeed, M admits supporting at \bar{x}_+ hyperplane Π ; as it was explained in the proof of (i), the set $\Pi \cap M$ (which clearly is convex, closed and bounded) is of affine dimension less than that one of M ; by the inductive hypothesis, the point \bar{x}_+ of this set is a convex combination of extreme points of the set, and by Lemma B.2.4 all these extreme points are extreme points of M as well. Thus, \bar{x}_+ is a convex combination of extreme points of M . Similar reasoning is valid for \bar{x}_- . ■

B.2.8.3 B.2.8.C. Example: Extreme points of a polyhedral set.

Consider a polyhedral set

$$K = \{x \in \mathbf{R}^n : Ax \leq b\},$$

A being a $m \times n$ matrix and b being a vector from \mathbf{R}^m . What are the extreme points of K ? The answer is given by the following

Theorem B.2.11 [Extreme points of polyhedral set]

Let $x \in K$. The vector x is an extreme point of K if and only if some n linearly independent (i.e., with linearly independent vectors of coefficients) inequalities of the system $Ax \leq b$ are equalities at x .

Proof. Let a_i^T , $i = 1, \dots, m$, be the rows of A .

The “only if” part: let x be an extreme point of K , and let I be the set of those indices i for which $a_i^T x = b_i$; we should prove that the set F of vectors $\{a_i : i \in I\}$ contains n linearly independent vectors, or, which is the same, that $\text{Lin}(F) = \mathbf{R}^n$. Assume that it is not the case; then the orthogonal complement to F contains a nonzero vector h (since the dimension of F^\perp is equal to $n - \dim \text{Lin}(F)$ and is therefore positive). Consider the segment $\Delta_\epsilon = [x - \epsilon h, x + \epsilon h]$, $\epsilon > 0$ being the parameter of our construction. Since h is orthogonal to the “active” vectors a_i

– those with $i \in I$, all points y of this segment satisfy the relations $a_i^T y = a_i^T x = b_i$. Now, if i is a “nonactive” index – one with $a_i^T x < b_i$ – then $a_i^T y \leq b_i$ for all $y \in \Delta_\epsilon$, provided that ϵ is small enough. Since there are finitely many nonactive indices, we can choose $\epsilon > 0$ in such a way that all $y \in \Delta_\epsilon$ will satisfy all “nonactive” inequalities $a_i^T y \leq b_i$, $i \notin I$. Since $y \in \Delta_\epsilon$ satisfies, as we have seen, also all “active” inequalities, we conclude that with the above choice of ϵ we get $\Delta_\epsilon \subset K$, which is a contradiction: $\epsilon > 0$ and $h \neq 0$, so that Δ_ϵ is a nontrivial segment with the midpoint x , and no such segment can be contained in K , since x is an extreme point of K . \square

To prove the “if” part, assume that $x \in K$ is such that among the inequalities $a_i^T x \leq b_i$ which are equalities at x there are n linearly independent, say, those with indices $1, \dots, n$, and let us prove that x is an extreme point of K . This is immediate: assuming that x is not an extreme point, we would get the existence of a nonzero vector h such that $x \pm h \in K$. In other words, for $i = 1, \dots, n$ we would have $b_i \pm a_i^T h \equiv a_i^T(x \pm h) \leq b_i$, which is possible only if $a_i^T h = 0$, $i = 1, \dots, n$. But the only vector which is orthogonal to n linearly independent vectors in \mathbf{R}^n is the zero vector (why?), and we get $h = 0$, which was assumed not to be the case. ■

Corollary B.2.1 *The set of extreme points of a polyhedral set is finite.*

Indeed, according to the above Theorem, every extreme point of a polyhedral set $K = \{x \in \mathbf{R}^n : Ax \leq b\}$ satisfies the equality version of certain n -inequality subsystem of the original system, the matrix of the subsystem being nonsingular. Due to the latter fact, an extreme point is uniquely defined by the corresponding subsystem, so that the number of extreme points does not exceed the number C_m^n of $n \times n$ submatrices of the matrix A and is therefore finite. ■

Note that C_m^n is nothing but an upper (and typically very conservative) bound on the number of extreme points of a polyhedral set given by m inequalities in \mathbf{R}^n : some $n \times n$ submatrices of A can be singular and, what is more important, the majority of the nonsingular ones normally produce “candidates” which do not satisfy the remaining inequalities.

Remark B.2.1 The result of Theorem B.2.11 is very important, in particular, for the theory of the Simplex method – the traditional computational tool of Linear Programming. When applied to the LP program in the standard form

$$\min_x \left\{ c^T x : Px = p, x \geq 0 \right\} \quad [x \in \mathbf{R}^n],$$

with $k \times n$ matrix P , the result of Theorem B.2.11 is that extreme points of the feasible set are exactly the *basic feasible solutions* of the system $Px = p$, i.e., nonnegative vectors x such that $Px = p$ and the set of columns of P associated with positive entries of x is linearly independent. Since the feasible set of an LP program in the standard form clearly does not contain lines, among the optimal solutions (if they exist) to an LP program in the standard form at least one is an extreme point of the feasible set (Theorem B.2.14.(ii)). Thus, in principle we could look through the finite set of all extreme points of the feasible set (\equiv through all basic feasible solutions) and to choose the one with the best value of the objective. This recipe allows to find a feasible solution in finitely many arithmetic operations, provided that the program is solvable, and is, basically, what the Simplex method does; this latter method, of course, looks through the basic feasible solutions in a smart way which normally allows to deal with a negligible part of them only.

Another useful consequence of Theorem B.2.11 is that if all the data in an LP program are rational, then every extreme point of the feasible domain of the program is a vector with rational entries. In particular, a solvable standard form LP program with rational data has at least one rational optimal solution.

B.2.8.C.1. Illustration: Birkhoff Theorem An $n \times n$ matrix X is called *double stochastic*, if its entries are nonnegative and all column and row sums are equal to 1. These matrices (treated as elements of $\mathbf{R}^{n^2} = \mathbf{R}^{n \times n}$) form a bounded polyhedral set, specifically, the set

$$\Pi_n = \{X = [x_{ij}]_{i,j=1}^n : x_{ij} \geq 0 \forall i, j, \sum_i x_{ij} = 1 \forall j, \sum_j x_{ij} = 1 \forall i\}$$

By Krein-Milman Theorem, Π_n is the convex hull of its extreme points. What are these extreme points? The answer is given by important

Theorem B.2.12 [Birkhoff Theorem] *Extreme points of Π_n are exactly the permutation matrices of order n , i.e., $n \times n$ Boolean (i.e., with 0/1 entries) matrices with exactly one nonzero element (equal to 1) in every row and every column.*

Exercise B.19 [Easy part] *Prove the easy part of the Theorem, specifically, that every $n \times n$ permutation matrix is an extreme point of Π_n .*

Proof of difficult part. Now let us prove that every extreme point of Π_n is a permutation matrix. To this end let us note that the $2n$ linear equations in the definition of Π_n — those saying that all row and column sums are equal to 1 — are linearly dependent, and dropping one of them, say, $\sum_i x_{in} = 1$, we do not alter the set. Indeed, the remaining equalities say that all row sums are equal to 1, so that the total sum of all entries in X is n , and that the first $n - 1$ column sums are equal to 1, meaning that the last column sum is $n - (n - 1) = 1$. Thus, we lose nothing when assuming that there are just $2n - 1$ equality constraints in the description of Π_n . Now let us prove the claim by induction in n . The base $n = 1$ is trivial. Let us justify the inductive step $n - 1 \Rightarrow n$. Thus, let X be an extreme point of Π_n . By Theorem B.2.11, among the constraints defining Π_n (i.e., $2n - 1$ equalities and n^2 inequalities $x_{ij} \geq 0$) there should be n^2 linearly independent which are satisfied at X as equations. Thus, at least $n^2 - (2n - 1) = (n - 1)^2$ entries in X should be zeros. It follows that at least one of the columns of X contains ≤ 1 nonzero entries (since otherwise the number of zero entries in X would be at most $n(n - 2) < (n - 1)^2$). Thus, there exists at least one column with at most 1 nonzero entry; since the sum of entries in this column is 1, this nonzero entry, let it be $x_{i\bar{j}}$, is equal to 1. Since the entries in row \bar{i} are nonnegative, sum up to 1 and $x_{i\bar{j}} = 1$, $x_{i\bar{j}} = 1$ is the only nonzero entry in its row and its column. Eliminating from X the row \bar{i} and the column \bar{j} , we get an $(n - 1) \times (n - 1)$ double stochastic matrix. By inductive hypothesis, this matrix is a convex combination of $(n - 1) \times (n - 1)$ permutation matrices. Augmenting every one of these matrices by the column and the row we have eliminated, we get a representation of X as a convex combination of $n \times n$ permutation matrices: $X = \sum_\ell \lambda_\ell P_\ell$ with nonnegative λ_ℓ summing up to 1. Since $P_\ell \in \Pi_n$ and X is an extreme point of Π_n , in this representation all terms with nonzero coefficients λ_ℓ must be equal to $\lambda_\ell X$, so that X is one of the permutation matrices P_ℓ and as such is a permutation matrix. \square

B.2.9 Structure of polyhedral sets

B.2.9.1 B.2.9.A. Main result

By definition, a polyhedral set M is the set of all solutions to a finite system of nonstrict linear inequalities:

$$M = \{x \in \mathbf{R}^n : Ax \leq b\}, \tag{B.2.18}$$

where A is a matrix of the column size n and certain row size m and b is m -dimensional vector. This is an “outer” description of a polyhedral set. We are about to establish an important result on the equivalent “inner” representation of a polyhedral set.

Consider the following construction. Let us take two finite nonempty sets of vectors V (“vertices”) and R (“rays”) and build the set

$$M(V, R) = \text{Conv}(V) + \text{Cone}(R) = \left\{ \sum_{v \in V} \lambda_v v + \sum_{r \in R} \mu_r r : \lambda_v \geq 0, \mu_r \geq 0, \sum_v \lambda_v = 1 \right\}.$$

Thus, we take all vectors which can be represented as sums of convex combinations of the points from V and conic combinations of the points from R . The set $M(V, R)$ clearly is convex (as the arithmetic sum of two convex sets $\text{Conv}(V)$ and $\text{Cone}(R)$). The promised inner description of polyhedral sets is as follows:

Theorem B.2.13 [Inner description of a polyhedral set] *The sets of the form $M(V, R)$ are exactly the nonempty polyhedral sets: $M(V, R)$ is polyhedral, and every nonempty polyhedral set M is $M(V, R)$ for properly chosen V and R .*

The polytopes $M(V, \{0\}) = \text{Conv}(V)$ are exactly the nonempty and bounded polyhedral sets. The sets of the type $M(\{0\}, R)$ are exactly the polyhedral cones (sets given by finitely many nonstrict homogeneous linear inequalities).

Remark B.2.2 In addition to the results of the Theorem, it can be proved that in the representation of a nonempty polyhedral set M as $M = \text{Conv}(V) + \text{Cone}(R)$

- the “conic” part $\text{Conv}(R)$ (not the set R itself!) is uniquely defined by M and is the recessive cone of M (see Comment to Lemma B.2.3);
- if M does not contain lines, then V can be chosen as the set of all extreme points of M .

Postponing temporarily the proof of Theorem B.2.13, let us explain why this theorem is that important – why it is so nice to know both inner and outer descriptions of a polyhedral set.

Consider a number of natural questions:

- A. Is it true that the inverse image of a polyhedral set $M \subset \mathbf{R}^n$ under an affine mapping $y \mapsto \mathcal{P}(y) = Py + p : \mathbf{R}^m \rightarrow \mathbf{R}^n$, i.e., the set

$$\mathcal{P}^{-1}(M) = \{y \in \mathbf{R}^m : Py + p \in M\}$$

is polyhedral?

- B. Is it true that the image of a polyhedral set $M \subset \mathbf{R}^n$ under an affine mapping $x \mapsto y = \mathcal{P}(x) = Px + p : \mathbf{R}^n \rightarrow \mathbf{R}^m$ – the set

$$\mathcal{P}(M) = \{Px + p : x \in M\}$$

is polyhedral?

- C. Is it true that the intersection of two polyhedral sets is again a polyhedral set?
- D. Is it true that the arithmetic sum of two polyhedral sets is again a polyhedral set?

The answers to all these questions are positive; one way to see it is to use calculus of polyhedral representations along with the fact that polyhedrally representable sets are exactly the same as polyhedral sets (section B.2.4). Another very instructive way is to use the just outlined results on the structure of polyhedral sets, which we intend to do now.

It is very easy to answer affirmatively to A, starting from the original – outer – definition of a polyhedral set: if $M = \{x : Ax \leq b\}$, then, of course,

$$\mathcal{P}^{-1}(M) = \{y : A(Py + p) \leq b\} = \{y : (AP)y \leq b - Ap\}$$

and therefore $\mathcal{P}^{-1}(M)$ is a polyhedral set.

An attempt to answer affirmatively to B via the same definition fails – there is no easy way to convert the linear inequalities defining a polyhedral set into those defining its image, and it is absolutely unclear why the image indeed is given by finitely many linear inequalities. Note, however, that there is no difficulty to answer affirmatively to B with the inner description of a nonempty polyhedral set: if $M = M(V, R)$, then, evidently,

$$\mathcal{P}(M) = M(\mathcal{P}(V), PR),$$

where $PR = \{Pr : r \in R\}$ is the image of R under the action of the homogeneous part of \mathcal{P} .

Similarly, positive answer to C becomes evident, when we use the outer description of a polyhedral set: taking intersection of the solution sets to two systems of nonstrict linear inequalities, we, of course, again get the solution set to a system of this type – you simply should put together all inequalities from the original two systems. And it is very unclear how to answer positively to D with the outer definition of a polyhedral set – what happens with inequalities when we add the solution sets? In contrast to this, the inner description gives the answer immediately:

$$\begin{aligned} M(V, R) + M(V', R') &= \text{Conv}(V) + \text{Cone}(R) + \text{Conv}(V') + \text{Cone}(R') \\ &= [\text{Conv}(V) + \text{Conv}(V')] + [\text{Cone}(R) + \text{Cone}(R')] \\ &= \text{Conv}(V + V') + \text{Cone}(R \cup R') \\ &= M(V + V', R \cup R'). \end{aligned}$$

Note that in this computation we used two rules which should be justified: $\text{Conv}(V) + \text{Conv}(V') = \text{Conv}(V + V')$ and $\text{Cone}(R) + \text{Cone}(R') = \text{Cone}(R \cup R')$. The second is evident from the definition of the conic hull, and only the first needs simple reasoning. To prove it, note that $\text{Conv}(V) + \text{Conv}(V')$ is a convex set which contains $V + V'$ and therefore contains $\text{Conv}(V + V')$. The inverse inclusion is proved as follows: if

$$x = \sum_i \lambda_i v_i, \quad y = \sum_j \lambda'_j v'_j$$

are convex combinations of points from V , resp., V' , then, as it is immediately seen (please check!),

$$x + y = \sum_{i,j} \lambda_i \lambda'_j (v_i + v'_j)$$

and the right hand side is a convex combination of points from $V + V'$.

We see that it is extremely useful to keep in mind both descriptions of polyhedral sets – what is difficult to see with one of them, is absolutely clear with another.

As a seemingly “more important” application of the developed theory, let us look at Linear Programming.

B.2.9.2 B.2.9.B. Theory of Linear Programming

A general Linear Programming program is the problem of maximizing a linear objective function over a polyhedral set:

$$(P) \quad \max_x \left\{ c^T x : x \in M = \{x \in \mathbf{R}^n : Ax \leq b\} \right\};$$

here c is a given n -dimensional vector – the objective, A is a given $m \times n$ constraint matrix and $b \in \mathbf{R}^m$ is the right hand side vector. Note that (P) is called “Linear Programming program in the canonical form”; there are other equivalent forms of the problem.

B.2.9.B.1. Solvability of a Linear Programming program. According to the Linear Programming terminology which you for sure know, (P) is called

- feasible, if it admits a feasible solution, i.e., the system $Ax \leq b$ is solvable, and infeasible otherwise;
- bounded, if it is feasible and the objective is above bounded on the feasible set, and unbounded, if it is feasible, but the objective is not bounded from above on the feasible set;
- solvable, if it is feasible and the optimal solution exists – the objective attains its maximum on the feasible set.

If the program is bounded, then the upper bound of the values of the objective on the feasible set is a real; this real is called the optimal value of the program and is denoted by c^* . It is convenient to assign optimal value to unbounded and infeasible programs as well – for an unbounded program it, by definition, is $+\infty$, and for an infeasible one it is $-\infty$.

Note that our terminology is aimed to deal with maximization programs; if the program is to minimize the objective, the terminology is updated in the natural way: when defining bounded/unbounded programs, we should speak about below boundedness rather than about the above boundedness of the objective, etc. E.g., the optimal value of an unbounded minimization program is $-\infty$, and of an infeasible one it is $+\infty$. This terminology is consistent with the usual way of converting a minimization problem into an equivalent maximization one by replacing the original objective c with $-c$: the properties of feasibility – boundedness – solvability remain unchanged, and the optimal value in all cases changes its sign.

I have said that you for sure know the above terminology; this is not exactly true, since you definitely have heard and used the words “infeasible LP program”, “unbounded LP program”, but hardly used the words “bounded LP program” – only the “solvable” one. This indeed is true, although absolutely unclear in advance – a bounded LP program always is solvable. We have already established this fact, even twice — via Fourier-Motzkin elimination (section B.2.4 and via the LP Duality Theorem). Let us reestablish this fundamental for Linear Programming fact with the tools we have at our disposal now.

Theorem B.2.14 (i) *A Linear Programming program is solvable if and only if it is bounded.*
(ii) *If the program is solvable and the feasible set of the program does not contain lines, then at least one of the optimal solutions is an extreme point of the feasible set.*

Proof. (i): The “only if” part of the statement is tautological: the definition of solvability includes boundedness. What we should prove is the “if” part – that a bounded program is

solvable. This is immediately given by the inner description of the feasible set M of the program: this is a polyhedral set, so that being nonempty (as it is for a bounded program), it can be represented as

$$M(V, R) = \text{Conv}(V) + \text{Cone}(R)$$

for some nonempty finite sets V and R . I claim first of all that since (P) is bounded, the inner product of c with every vector from R is nonpositive. Indeed, otherwise there would be $r \in R$ with $c^T r > 0$; since $M(V, R)$ clearly contains with every its point x the entire ray $\{x + tr : t \geq 0\}$, and the objective evidently is unbounded on this ray, it would be above unbounded on M , which is not the case.

Now let us choose in the *finite and nonempty* set V the point, let it be called v^* , which maximizes the objective on V . I claim that v^* is an optimal solution to (P), so that (P) is solvable. The justification of the claim is immediate: v^* clearly belongs to M ; now, a generic point of $M = M(V, R)$ is

$$x = \sum_{v \in V} \lambda_v v + \sum_{r \in R} \mu_r r$$

with nonnegative λ_v and μ_r and with $\sum_v \lambda_v = 1$, so that

$$\begin{aligned} c^T x &= \sum_v \lambda_v c^T v + \sum_r \mu_r c^T r \\ &\leq \sum_v \lambda_v c^T v && [\text{since } \mu_r \geq 0 \text{ and } c^T r \leq 0, r \in R] \\ &\leq \sum_v \lambda_v c^T v^* && [\text{since } \lambda_v \geq 0 \text{ and } c^T v \leq c^T v^*] \\ &= c^T v^* && [\text{since } \sum_v \lambda_v = 1] \square \end{aligned}$$

(ii): if the feasible set of (P), let it be called M , does not contain lines, it, being convex and closed (as a polyhedral set) possesses extreme points. It follows that (ii) is valid in the trivial case when the objective of (ii) is constant on the entire feasible set, since then every extreme point of M can be taken as the desired optimal solution. The case when the objective is nonconstant on M can be immediately reduced to the aforementioned trivial case: if x^* is an optimal solution to (P) and the linear form $c^T x$ is nonconstant on M , then the hyperplane $\Pi = \{x : c^T x = c^* \}$ is supporting to M at x^* ; the set $\Pi \cap M$ is closed, convex, nonempty and does not contain lines, therefore it possesses an extreme point x^{**} which, on one hand, clearly is an optimal solution to (P), and on another hand is an extreme point of M by Lemma B.2.4. ■

B.2.9.3 B.2.9.C. Structure of a polyhedral set: proofs

B.2.9.C.1. Structure of a bounded polyhedral set. Let us start with proving a significant part of Theorem B.2.13 – the one describing *bounded* polyhedral sets.

Theorem B.2.15 [Structure of a bounded polyhedral set] *A bounded and nonempty polyhedral set M in \mathbf{R}^n is a polytope, i.e., is the convex hull of a finite nonempty set:*

$$M = M(V, \{0\}) = \text{Conv}(V);$$

one can choose as V the set of all extreme points of M .

Vice versa – a polytope is a bounded and nonempty polyhedral set.

Proof. The first part of the statement – that a bounded nonempty polyhedral set is a polytope – is readily given by the Krein-Milman Theorem combined with Corollary B.2.1. Indeed, a polyhedral set always is closed (as a set given by nonstrict inequalities involving continuous functions) and convex; if it is also bounded and nonempty, it, by the Krein-Milman Theorem, is the convex hull of the set V of its extreme points; V is finite by Corollary B.2.1. \square

Now let us prove the more difficult part of the statement – that a polytope is a bounded polyhedral set. The fact that a convex hull of a finite set is bounded is evident. Thus, all we need is to prove that the convex hull of finitely many points is a polyhedral set. To this end note that this convex hull clearly is polyhedrally representable:

$$\text{Conv}\{v_1, \dots, v_N\} = \{x : \exists \lambda : \lambda \geq 0, \sum_i \lambda_i = 1, x = \sum_i \lambda_i v_i\}$$

and therefore is polyhedral by Theorem B.2.5. ■

B.2.9.C.2. Structure of a general polyhedral set: completing the proof. Now let us prove the general Theorem B.2.13. The proof basically follows the lines of the one of Theorem B.2.15, but with one elaboration: now we cannot use the Krein-Milman Theorem to take upon itself part of our difficulties.

To simplify language let us call VR-sets (“V” from “vertex”, “R” from rays) the sets of the form $M(V, R)$, and P-sets the nonempty polyhedral sets. We should prove that every P-set is a VR-set, and vice versa. We start with proving that every P-set is a VR-set.

B.2.9.C.2.A. $P \Rightarrow VR$:

$P \Rightarrow VR$, Step 1: reduction to the case when the P-set does not contain lines. Let M be a P-set, so that M is the set of all solutions to a solvable system of linear inequalities:

$$M = \{x \in \mathbf{R}^n : Ax \leq b\} \quad (\text{B.2.19})$$

with $m \times n$ matrix A . Such a set may contain lines; if h is the direction of a line in M , then $A(x + th) \leq b$ for some x and all $t \in \mathbf{R}$, which is possible only if $Ah = 0$. Vice versa, if h is from the kernel of A , i.e., if $Ah = 0$, then the line $x + \mathbf{R}^h$ with $x \in M$ clearly is contained in M . Thus, we come to the following fact:

Lemma B.2.5 *Nonempty polyhedral set (B.2.19) contains lines if and only if the kernel of A is nontrivial, and the nonzero vectors from the kernel are exactly the directions of lines contained in M : if M contains a line with direction h , then $h \in \text{Ker } A$, and vice versa: if $0 \neq h \in \text{Ker } A$ and $x \in M$, then M contains the entire line $x + \mathbf{R}h$.*

Given a nonempty set (B.2.19), let us denote by L the kernel of A and by L^\perp the orthogonal complement to the kernel, and let M' be the cross-section of M by L^\perp :

$$M' = \{x \in L^\perp : Ax \leq b\}.$$

The set M' clearly does not contain lines (since the direction of every line in M' , on one hand, should belong to L^\perp due to $M' \subset L^\perp$, and on the other hand – should belong to $L = \text{Ker } A$, since a line in $M' \subset M$ is a line in M as well). The set M' is nonempty and, moreover, $M = M' + L$.

Indeed, M' contains the orthogonal projections of all points from M onto L^\perp (since to project a point onto L^\perp , you should move from this point along certain line with the direction in L , and all these movements, started in M , keep you in M by the Lemma) and therefore is nonempty, first, and is such that $M' + L \supset M$, second. On the other hand, $M' \subset M$ and $M + L = M$ by Lemma B.2.5, whence $M' + L \subset M$. Thus, $M' + L = M$.

Finally, M' is a polyhedral set together with M , since the inclusion $x \in L^\perp$ can be represented by $\dim L$ linear equations (i.e., by $2\dim L$ nonstrict linear inequalities): you should say that x is orthogonal to $\dim L$ somehow chosen vectors $a_1, \dots, a_{\dim L}$ forming a basis in L .

The results of our effort are as follows: given an arbitrary P-set M , we have represented it as the sum of a P-set M' not containing lines and a linear subspace L . With this decomposition in mind we see that in order to achieve our current goal – to prove that every P-set is a VR-set – it suffices to prove the same statement for P-sets not containing lines. Indeed, given that $M' = M(V, R')$ and denoting by R' a finite set such that $L = \text{Cone}(R')$ (to get R' , take the set of $2\dim L$ vectors $\pm a_i$, $i = 1, \dots, \dim L$, where $a_1, \dots, a_{\dim L}$ is a basis in L), we would obtain

$$\begin{aligned} M &= M' + L \\ &= [\text{Conv}(V) + \text{Cone}(R)] + \text{Cone}(R') \\ &= \text{Conv}(V) + [\text{Cone}(R) + \text{Cone}(R')] \\ &= \text{Conv}(V) + \text{Cone}(R \cup R') \\ &= M(V, R \cup R') \end{aligned}$$

We see that in order to establish that a P-set is a VR-set it suffices to prove the same statement for the case when the P-set in question does not contain lines.

P⇒VR, Step 2: the P-set does not contain lines. Our situation is as follows: we are given a not containing lines P-set in \mathbf{R}^n and should prove that it is a VR-set. We shall prove this statement by induction on the dimension n of the space. The case of $n = 0$ is trivial. Now assume that the statement in question is valid for $n \leq k$, and let us prove that it is valid also for $n = k + 1$. Let M be a not containing lines P-set in \mathbf{R}^{k+1} :

$$M = \{x \in \mathbf{R}^{k+1} : a_i^T x \leq b_i, i = 1, \dots, m\}. \quad (\text{B.2.20})$$

Without loss of generality we may assume that all a_i are nonzero vectors (since M is nonempty, the inequalities with $a_i = 0$ are valid on the entire \mathbf{R}^n , and removing them from the system, we do not vary its solution set). Note that $m > 0$ – otherwise M would contain lines, since $k \geq 0$.

¹⁰. We may assume that M is unbounded – otherwise the desired result is given already by Theorem B.2.15. By Exercise B.18, there exists a recessive direction $r \neq 0$ of M . Thus, M contains the ray $\{x + tr : t \geq 0\}$, whence, by Lemma B.2.3, $M + \text{Cone}(\{r\}) = M$. \square

²⁰. For every $i \leq m$, where m is the row size of the matrix A from (B.2.20), that is, the number of linear inequalities in the description of M , let us denote by M_i the corresponding “facet” of M – the polyhedral set given by the system of inequalities (B.2.20) with the inequality $a_i^T x \leq b_i$ replaced by the equality $a_i^T x = b_i$. Some of these “facets” can be empty; let I be the set of indices i of nonempty M_i ’s.

When $i \in I$, the set M_i is a nonempty polyhedral set – i.e., a P-set – which does not contain lines (since $M_i \subset M$ and M does not contain lines). Besides this, M_i belongs to the hyperplane $\{a_i^T x = b_i\}$, i.e., actually it is a P-set in \mathbf{R}^k . By the inductive hypothesis, we have representations

$$M_i = M(V_i, R_i), \quad i \in I,$$

for properly chosen finite nonempty sets V_i and R_i . I claim that

$$M = M(\cup_{i \in I} V_i, \cup_{i \in I} R_i \cup \{r\}), \quad (\text{B.2.21})$$

where r is a recessive direction of M found in 1⁰; after the claim will be supported, our induction will be completed.

To prove (B.2.21), note, first of all, that the right hand side of this relation is contained in the left hand side one. Indeed, since $M_i \subset M$ and $V_i \subset M_i$, we have $V_i \subset M$, whence also $V = \cup_i V_i \subset M$; since M is convex, we have

$$\text{Conv}(V) \subset M. \quad (\text{B.2.22})$$

Further, if $r' \in R_i$, then r' is a recessive direction of M_i ; since $M_i \subset M$, r' is a recessive direction of M by Lemma B.2.3. Thus, every vector from $\cup_{i \in I} R_i$ is a recessive direction for M , same as r ; thus, every vector from $R = \cup_{i \in I} R_i \cup \{r\}$ is a recessive direction of M , whence, again by Lemma B.2.3,

$$M + \text{Cone}(R) = M.$$

Combining this relation with (B.2.22), we get $M(V, R) \subset M$, as claimed.

It remains to prove that M is contained in the right hand side of (B.2.21). Let $x \in M$, and let us move from x along the direction $(-r)$, i.e., move along the ray $\{x - tr : t \geq 0\}$. After large enough step along this ray we leave M . (Indeed, otherwise the ray with the direction $-r$ started at x would be contained in M , while the opposite ray for sure is contained in M since r is a recessive direction of M ; we would conclude that M contains a line, which is not the case by assumption.) Since the ray $\{x - tr : t \geq 0\}$ eventually leaves M and M is bounded, there exists the largest t , let it be called t^* , such that $x' = x - t^*r$ still belongs to M . It is clear that at x' one of the linear inequalities defining M becomes equality – otherwise we could slightly increase the parameter t^* still staying in M . Thus, $x' \in M_i$ for some $i \in I$. Consequently,

$$x' \in \text{Conv}(V_i) + \text{Cone}(R_i),$$

whence $x = x' + t^*r \in \text{Conv}(V_i) + \text{Cone}(R_i \cup \{r\}) \subset M(V, R)$, as claimed. \square

B.2.9.C.2.B. VR \Rightarrow P: We already know that every P-set is a VR-set. Now we shall prove that every VR-set is a P-set, thus completing the proof of Theorem B.2.13. This is immediate: a VR-set is polyhedrally representable (why?) and thus is a P-set by Theorem B.2.5. ■

Appendix C

Convex functions

C.1 Convex functions: first acquaintance

C.1.1 Definition and Examples

Definition C.1.1 [convex function] A function $f : Q \rightarrow \mathbf{R}$ defined on a nonempty subset Q of \mathbf{R}^n and taking real values is called convex, if

- the domain Q of the function is convex;
- for every $x, y \in Q$ and every $\lambda \in [0, 1]$ one has

$$f(\lambda x + (1 - \lambda)y) \leq \lambda f(x) + (1 - \lambda)f(y). \quad (\text{C.1.1})$$

If the above inequality is strict whenever $x \neq y$ and $0 < \lambda < 1$, f is called strictly convex.

A function f such that $-f$ is convex is called concave; the domain Q of a concave function should be convex, and the function itself should satisfy the inequality opposite to (C.1.1):

$$f(\lambda x + (1 - \lambda)y) \geq \lambda f(x) + (1 - \lambda)f(y), \quad x, y \in Q, \lambda \in [0, 1].$$

The simplest example of a convex function is an *affine function*

$$f(x) = a^T x + b$$

– the sum of a linear form and a constant. This function clearly is convex on the entire space, and the “convexity inequality” for it is equality. An affine function is both convex and concave; it is easily seen that a function which is both convex and concave on the entire space is affine.

Here are several elementary examples of “nonlinear” convex functions of one variable:

- functions convex on the whole axis:

x^{2p} , p is a positive integer;

$\exp\{x\}$;

- functions convex on the nonnegative ray:

x^p , $1 \leq p$;

$-x^p$, $0 \leq p \leq 1$;

$x \ln x$;

- functions convex on the positive ray:

$$\begin{aligned} 1/x^p, p > 0; \\ -\ln x. \end{aligned}$$

To the moment it is not clear why these functions are convex; in the mean time we shall derive a simple analytic criterion for detecting convexity which immediately demonstrates that the above functions indeed are convex.

A very convenient equivalent definition of a convex function is in terms of its *epigraph*. Given a real-valued function f defined on a nonempty subset Q of \mathbf{R}^n , we define its epigraph as the set

$$\text{Epi}(f) = \{(t, x) \in \mathbf{R}^{n+1} : x \in Q, t \geq f(x)\};$$

geometrically, to define the epigraph, you should take the *graph* of the function – the surface $\{t = f(x), x \in Q\}$ in \mathbf{R}^{n+1} – and add to this surface all points which are “above” it. The equivalent, more geometrical, definition of a convex function is given by the following simple statement (prove it!):

Proposition C.1.1 [definition of convexity in terms of the epigraph]

A function f defined on a subset of \mathbf{R}^n is convex if and only if its epigraph is a nonempty convex set in \mathbf{R}^{n+1} .

More examples of convex functions: norms. Equipped with Proposition C.1.1, we can extend our initial list of convex functions (several one-dimensional functions and affine ones) by more examples – *norms*. Let $\pi(x)$ be a norm on \mathbf{R}^n (see Section B.1.2.B). To the moment we know three examples of norms – the Euclidean norm $\|x\|_2 = \sqrt{x^T x}$, the 1-norm $\|x\|_1 = \sum_i |x_i|$ and the ∞ -norm $\|x\|_\infty = \max_i |x_i|$. It was also claimed (although not proved) that these are three members of an infinite family of norms

$$\|x\|_p = \left(\sum_{i=1}^n |x_i|^p \right)^{1/p}, \quad 1 \leq p \leq \infty$$

(the right hand side of the latter relation for $p = \infty$ is, by definition, $\max_i |x_i|$).

We are about to prove that every norm is convex:

Proposition C.1.2 *Let $\pi(x)$ be a real-valued function on \mathbf{R}^n which is positively homogeneous of degree 1:*

$$\pi(tx) = t\pi(x) \quad \forall x \in \mathbf{R}^n, t \geq 0.$$

π is convex if and only if it is subadditive:

$$\pi(x + y) \leq \pi(x) + \pi(y) \quad \forall x, y \in \mathbf{R}^n.$$

In particular, a norm (which by definition is positively homogeneous of degree 1 and is subadditive) is convex.

Proof is immediate: the epigraph of a positively homogeneous of degree 1 function π clearly is a conic set: $(t, x) \in \text{Epi}(\pi) \Rightarrow \lambda(t, x) \in \text{Epi}(\pi)$ whenever $\lambda \geq 0$. Now, by Proposition C.1.1 π is convex if and only if $\text{Epi}(\pi)$ is convex. It is clear that a conic set is convex if and only if it contains the sum of every two its elements (why ?); this latter property is satisfied for the epigraph of a real-valued function if and only if the function is subadditive (evident). ■

C.1.2 Elementary properties of convex functions

C.1.2.1 C.1.2.A. Jensen's inequality

The following elementary observation is, I believe, one of the most useful observations in the world:

Proposition C.1.3 [Jensen's inequality] *Let f be convex and Q be the domain of f . Then for every convex combination*

$$\sum_{i=1}^N \lambda_i x_i$$

of points from Q one has

$$f\left(\sum_{i=1}^N \lambda_i x_i\right) \leq \sum_{i=1}^N \lambda_i f(x_i).$$

The proof is immediate: the points $(f(x_i), x_i)$ clearly belong to the epigraph of f ; since f is convex, its epigraph is a convex set, so that the convex combination

$$\sum_{i=1}^N \lambda_i (f(x_i), x_i) = \left(\sum_{i=1}^N \lambda_i f(x_i), \sum_{i=1}^N \lambda_i x_i\right)$$

of the points also belongs to $\text{Epi}(f)$. By definition of the epigraph, the latter means exactly that $\sum_{i=1}^N \lambda_i f(x_i) \geq f\left(\sum_{i=1}^N \lambda_i x_i\right)$. ■

Note that the definition of convexity of a function f is exactly the requirement on f to satisfy the Jensen inequality for the case of $N = 2$; we see that to satisfy this inequality for $N = 2$ is the same as to satisfy it for all N .

C.1.2.2 C.1.2.B. Convexity of level sets of a convex function

The following simple observation is also very useful:

Proposition C.1.4 [convexity of level sets] *Let f be a convex function with the domain Q . Then, for every real α , the set*

$$\text{lev}_\alpha(f) = \{x \in Q : f(x) \leq \alpha\}$$

– the level set of f – is convex.

The proof takes one line: if $x, y \in \text{lev}_\alpha(f)$ and $\lambda \in [0, 1]$, then $f(\lambda x + (1 - \lambda)y) \leq \lambda f(x) + (1 - \lambda)f(y) \leq \lambda\alpha + (1 - \lambda)\alpha = \alpha$, so that $\lambda x + (1 - \lambda)y \in \text{lev}_\alpha(f)$.

Note that the convexity of level sets does not characterize convex functions; there are non-convex functions which share this property (e.g., every monotone function on the axis). The “proper” characterization of convex functions in terms of convex sets is given by Proposition C.1.1 – convex functions are exactly the functions with convex epigraphs. Convexity of level sets specify a wider family of functions, the so called quasiconvex ones.

C.1.3 What is the value of a convex function outside its domain?

Literally, the question which entitles this subsection is senseless. Nevertheless, when speaking about convex functions, it is extremely convenient to think that the function outside its domain also has a value, namely, takes the value $+\infty$; with this convention, we can say that

a convex function f on \mathbf{R}^n is a function taking values in the extended real axis $\mathbf{R} \cup \{+\infty\}$ such that the domain $\text{Dom } f$ of the function – the set of those x 's where $f(x)$ is finite – is nonempty, and for all $x, y \in \mathbf{R}^n$ and all $\lambda \in [0, 1]$ one has

$$f(\lambda x + (1 - \lambda)y) \leq \lambda f(x) + (1 - \lambda)f(y). \quad (\text{C.1.2})$$

If the expression in the right hand side involves infinities, it is assigned the value according to the standard and reasonable conventions on what are arithmetic operations in the “extended real axis” $\mathbf{R} \cup \{+\infty\} \cup \{-\infty\}$:

- arithmetic operations with reals are understood in their usual sense;
- the sum of $+\infty$ and a real, same as the sum of $+\infty$ and $+\infty$ is $+\infty$; similarly, the sum of a real and $-\infty$, same as the sum of $-\infty$ and $-\infty$ is $-\infty$. The sum of $+\infty$ and $-\infty$ is undefined;
- the product of a real and $+\infty$ is $+\infty$, 0 or $-\infty$, depending on whether the real is positive, zero or negative, and similarly for the product of a real and $-\infty$. The product of two “infinities” is again infinity, with the usual rule for assigning the sign to the product.

Note that it is not clear in advance that our new definition of a convex function is equivalent to the initial one: initially we included into the definition requirement for the domain to be convex, and now we omit explicit indicating this requirement. In fact, of course, the definitions are equivalent: convexity of $\text{Dom } f$ – i.e., the set where f is finite – is an immediate consequence of the “convexity inequality” (C.1.2).

It is convenient to think of a convex function as of something which is defined everywhere, since it saves a lot of words. E.g., with this convention I can write $f + g$ (f and g are convex functions on \mathbf{R}^n), and everybody will understand what is meant; without this convention, I am supposed to add to this expression the explanation as follows: “ $f + g$ is a function with the domain being the intersection of those of f and g , and in this intersection it is defined as $(f + g)(x) = f(x) + g(x)$ ”.

C.2 How to detect convexity

In an optimization problem

$$\min_x \{f(x) : g_j(x) \leq 0, j = 1, \dots, m\}$$

convexity of the objective f and the constraints g_i is crucial: it turns out that problems with this property possess nice theoretical properties (e.g., the local *necessary optimality conditions* for these problems are *sufficient for global optimality*); and what is much more important, convex problems can be efficiently (both in theoretical and, to some extent, in the practical meaning of the word) solved numerically, which is not, unfortunately, the case for general nonconvex

problems. This is why it is so important to know how one can detect convexity of a given function. This is the issue we are coming to.

The scheme of our investigation is typical for mathematics. Let me start with the example which you know from Analysis. How do you detect continuity of a function? Of course, there is a definition of continuity in terms of ϵ and δ , but it would be an actual disaster if each time we need to prove continuity of a function, we were supposed to write down the proof that "for every positive ϵ there exists positive δ such that ...". In fact we use another approach: we list once for ever a number of standard operations which preserve continuity, like addition, multiplication, taking superpositions, etc., and point out a number of standard examples of continuous functions – like the power function, the exponent, etc. To prove that the operations in the list preserve continuity, same as to prove that the standard functions are continuous, this takes certain effort and indeed is done in $\epsilon - \delta$ terms; but after this effort is once invested, we normally have no difficulties with proving continuity of a given function: it suffices to demonstrate that the function can be obtained, in finitely many steps, from our "raw materials" – the standard functions which are known to be continuous – by applying our machinery – the combination rules which preserve continuity. Normally this demonstration is given by a single word "evident" or even is understood by default.

This is exactly the case with convexity. Here we also should point out the list of operations which preserve convexity and a number of standard convex functions.

C.2.1 Operations preserving convexity of functions

These operations are as follows:

- [stability under taking weighted sums] if f, g are convex functions on \mathbf{R}^n , then their linear combination $\lambda f + \mu g$ with nonnegative coefficients again is convex, provided that it is finite at least at one point;

[this is given by straightforward verification of the definition]

- [stability under affine substitutions of the argument] the superposition $f(Ax + b)$ of a convex function f on \mathbf{R}^n and affine mapping $x \mapsto Ax + b$ from \mathbf{R}^m into \mathbf{R}^n is convex, provided that it is finite at least at one point.

[you can prove it directly by verifying the definition or by noting that the epigraph of the superposition, if nonempty, is the inverse image of the epigraph of f under an affine mapping]

- [stability under taking pointwise sup] upper bound $\sup_{\alpha} f_{\alpha}(\cdot)$ of every family of convex functions on \mathbf{R}^n is convex, provided that this bound is finite at least at one point.

[to understand it, note that the epigraph of the upper bound clearly is the intersection of epigraphs of the functions from the family; recall that the intersection of every family of convex sets is convex]

- ["Convex Monotone superposition"] Let $f(x) = (f_1(x), \dots, f_k(x))$ be vector-function on \mathbf{R}^n with convex components f_i , and assume that F is a convex function on \mathbf{R}^k which is monotone, i.e., such that $z \leq z'$ always implies that $F(z) \leq F(z')$. Then the superposition

$$\phi(x) = F(f(x)) = F(f_1(x), \dots, f_k(x))$$

is convex on \mathbf{R}^n , provided that it is finite at least at one point.

Remark C.2.1 The expression $F(f_1(x), \dots, f_k(x))$ makes no evident sense at a point x where some of f_i 's are $+\infty$. By definition, we assign the superposition at such a point the value $+\infty$.

[To justify the rule, note that if $\lambda \in (0, 1)$ and $x, x' \in \text{Dom } \phi$, then $z = f(x), z' = f(x')$ are vectors from \mathbf{R}^k which belong to $\text{Dom } F$, and due to the convexity of the components of f we have

$$f(\lambda x + (1 - \lambda)x') \leq \lambda z + (1 - \lambda)z';$$

in particular, the left hand side is a vector from \mathbf{R}^k – it has no “infinite entries”, and we may further use the monotonicity of F :

$$\phi(\lambda x + (1 - \lambda)x') = F(f(\lambda x + (1 - \lambda)x')) \leq F(\lambda z + (1 - \lambda)z')$$

and now use the convexity of F :

$$F(\lambda z + (1 - \lambda)z') \leq \lambda F(z) + (1 - \lambda)F(z')$$

to get the required relation

$$\phi(\lambda x + (1 - \lambda)x') \leq \lambda\phi(x) + (1 - \lambda)\phi(x').$$

]

Imagine how many extra words would be necessary here if there were no convention on the value of a convex function outside its domain!

Two more rules are as follows:

- [stability under partial minimization] if $f(x, y) : \mathbf{R}_x^n \times \mathbf{R}_y^m$ is convex (as a function of $z = (x, y)$; this is called *joint convexity*) and the function

$$g(x) = \inf_y f(x, y)$$

is *proper*, i.e., is $> -\infty$ everywhere and is finite at least at one point, then g is convex [this can be proved as follows. We should prove that if $x, x' \in \text{Dom } g$ and $x'' = \lambda x + (1 - \lambda)x'$ with $\lambda \in [0, 1]$, then $x'' \in \text{Dom } g$ and $g(x'') \leq \lambda g(x) + (1 - \lambda)g(x')$. Given positive ϵ , we can find y and y' such that $(x, y) \in \text{Dom } f$, $(x', y') \in \text{Dom } f$ and $g(x) + \epsilon \geq f(x, y)$, $g(y') + \epsilon \geq f(x', y')$. Taking weighted sum of these two inequalities, we get

$$\lambda g(x) + (1 - \lambda)g(y) + \epsilon \geq \lambda f(x, y) + (1 - \lambda)f(x', y') \geq$$

[since f is convex]

$$\geq f(\lambda x + (1 - \lambda)x', \lambda y + (1 - \lambda)y') = f(x'', \lambda y + (1 - \lambda)y')$$

(the last \geq follows from the convexity of f). The concluding quantity in the chain is $\geq g(x'')$, and we get $g(x'') \leq \lambda g(x) + (1 - \lambda)g(x') + \epsilon$. In particular, $x'' \in \text{Dom } g$ (recall that g is assumed to take only the values from \mathbf{R} and the value $+\infty$). Moreover, since the resulting inequality is valid for all $\epsilon > 0$, we come to $g(x'') \leq \lambda g(x) + (1 - \lambda)g(x')$, as required.]

- the “conic transformation” of a convex function f on \mathbf{R}^n – the function $g(y, x) = yf(x/y)$ – is convex in the half-space $y > 0$ in \mathbf{R}^{n+1} .

Now we know what are the basic operations preserving convexity. Let us look what are the standard functions these operations can be applied to. A number of examples was already given, but we still do not know why the functions in the examples are convex. The usual way to check convexity of a “simple” – given by a simple formula – function is based on *differential criteria of convexity*. Let us look what are these criteria.

C.2.2 Differential criteria of convexity

From the definition of convexity of a function it immediately follows that convexity is one-dimensional property: a proper (i.e., finite at least at one point) function f on \mathbf{R}^n taking values in $\mathbf{R} \cup \{+\infty\}$ is convex if and only if its restriction on every line, i.e., every function of the type $g(t) = f(x + th)$ on the axis, is either convex, or is identically $+\infty$.

It follows that to detect convexity of a function, it, in principle, suffices to know how to detect convexity of functions of one variable. This latter question can be resolved by the standard Calculus tools. Namely, in the Calculus they prove the following simple

Proposition C.2.1 [Necessary and Sufficient Convexity Condition for smooth functions on the axis] *Let (a, b) be an interval in the axis (we do not exclude the case of $a = -\infty$ and/or $b = +\infty$). Then*

- (i) *A differentiable everywhere on (a, b) function f is convex on (a, b) if and only if its derivative f' is monotonically nondecreasing on (a, b) ;*
- (ii) *A twice differentiable everywhere on (a, b) function f is convex on (a, b) if and only if its second derivative f'' is nonnegative everywhere on (a, b) .*

With the Proposition, you can immediately verify that the functions listed as examples of convex functions in Section C.1.1 indeed are convex. The only difficulty which you may meet is that some of these functions (e.g., x^p , $p \geq 1$, and $-x^p$, $0 \leq p \leq 1$, were claimed to be convex on the half-interval $[0, +\infty)$, while the Proposition speaks about convexity of functions on intervals. To overcome this difficulty, you may use the following simple

Proposition C.2.2 *Let M be a convex set and f be a function with $\text{Dom } f = M$. Assume that f is convex on $\text{ri } M$ and is continuous on M , i.e.,*

$$f(x_i) \rightarrow f(x), i \rightarrow \infty,$$

whenever $x_i, x \in M$ and $x_i \rightarrow x$ as $i \rightarrow \infty$. Then f is convex on M .

Proof of Proposition C.2.1:

(i), necessity. Assume that f is differentiable and convex on (a, b) ; we should prove that then f' is monotonically nondecreasing. Let $x < y$ be two points of (a, b) , and let us prove that $f'(x) \leq f'(y)$. Indeed, let $z \in (x, y)$. We clearly have the following representation of z as a convex combination of x and y :

$$z = \frac{y-z}{y-x}x + \frac{x-z}{y-x}y,$$

whence, from convexity,

$$f(z) \leq \frac{y-z}{y-x}f(x) + \frac{x-z}{y-x}f(y),$$

whence

$$\frac{f(z) - f(x)}{x - z} \leq \frac{f(y) - f(z)}{y - z}.$$

Passing here to limit as $z \rightarrow x + 0$, we get

$$f'(x) \leq \frac{(f(y) - f(x))}{y - x},$$

and passing in the same inequality to limit as $z \rightarrow y - 0$, we get

$$f'(y) \geq \frac{(f(y) - f(x))}{y - x},$$

whence $f'(x) \leq f'(y)$, as claimed. ■

(i), sufficiency. We should prove that if f is differentiable on (a, b) and f' is monotonically nondecreasing on (a, b) , then f is convex on (a, b) . It suffices to verify that if $x < y$, $x, y \in (a, b)$, and $z = \lambda x + (1 - \lambda)y$ with $0 < \lambda < 1$, then

$$f(z) \leq \lambda f(x) + (1 - \lambda)f(y),$$

or, which is the same (write $f(z)$ as $\lambda f(z) + (1 - \lambda)f(z)$), that

$$\frac{f(z) - f(x)}{\lambda} \leq \frac{f(y) - f(z)}{1 - \lambda}.$$

noticing that $z - x = \lambda(y - x)$ and $y - z = (1 - \lambda)(y - x)$, we see that the inequality we should prove is equivalent to

$$\frac{f(z) - f(x)}{z - x} \leq \frac{f(y) - f(z)}{y - z}.$$

But in this equivalent form the inequality is evident: by the Lagrange Mean Value Theorem, its left hand side is $f'(\xi)$ with some $\xi \in (x, z)$, while the right hand one is $f'(\eta)$ with some $\eta \in (z, y)$. Since f' is nondecreasing and $\xi \leq z \leq \eta$, we have $f'(\xi) \leq f'(\eta)$, and the left hand side in the inequality we should prove indeed is \leq the right hand one. □

(ii) is immediate consequence of (i), since, as we know from the very beginning of Calculus, a differentiable function – in the case in question, it is f' – is monotonically nondecreasing on an interval if and only if its derivative is nonnegative on this interval. ■

In fact, for functions of one variable there is a differential criterion of convexity which does not assume any smoothness (we shall not prove this criterion):

Proposition C.2.3 [convexity criterion for univariate functions]

Let $g : \mathbf{R} \rightarrow \mathbf{R} \cup \{+\infty\}$ be a function. Let the domain $\Delta = \{t : g(t) < \infty\}$ of the function be a convex set which is not a singleton, i.e., let it be an interval (a, b) with possibly added one or both endpoints ($-\infty \leq a < b \leq \infty$). g is convex if and only if it satisfies the following 3 requirements:

- 1) g is continuous on (a, b) ;
- 2) g is differentiable everywhere on (a, b) , excluding, possibly, a countable set of points, and the derivative $g'(t)$ is nondecreasing on its domain;
- 3) at each endpoint u of the interval (a, b) which belongs to Δ g is upper semicontinuous:

$$g(u) \geq \limsup_{t \in (a, b), t \rightarrow u} g(t).$$

Proof of Proposition C.2.2: Let $x, y \in M$ and $z = \lambda x + (1 - \lambda)y$, $\lambda \in [0, 1]$, and let us prove that

$$f(z) \leq \lambda f(x) + (1 - \lambda)f(y).$$

As we know from Theorem B.1.1.(iii), there exist sequences $x_i \in \text{ri } M$ and $y_i \in \text{ri } M$ converging, respectively to x and to y . Then $z_i = \lambda x_i + (1 - \lambda)y_i$ converges to z as $i \rightarrow \infty$, and since f is convex on $\text{ri } M$, we have

$$f(z_i) \leq \lambda f(x_i) + (1 - \lambda)f(y_i);$$

passing to limit and taking into account that f is continuous on M and x_i, y_i, z_i converge, as $i \rightarrow \infty$, to $x, y, z \in M$, respectively, we obtain the required inequality. ■

From Propositions C.2.1.(ii) and C.2.2 we get the following convenient necessary and sufficient condition for convexity of a *smooth* function of n variables:

Corollary C.2.1 [convexity criterion for smooth functions on \mathbf{R}^n]

Let $f : \mathbf{R}^n \rightarrow \mathbf{R} \cup \{+\infty\}$ be a function. Assume that the domain Q of f is a convex set with a nonempty interior and that f is

- continuous on Q
and
- twice differentiable on the interior of Q .

Then f is convex if and only if its Hessian is positive semidefinite on the interior of Q :

$$h^T f''(x)h \geq 0 \quad \forall x \in \text{int } Q \quad \forall h \in \mathbf{R}^n.$$

Proof. The "only if" part is evident: if f is convex and $x \in Q' = \text{int } Q$, then the function of one variable

$$g(t) = f(x + th)$$

(h is an arbitrary fixed direction in \mathbf{R}^n) is convex in certain neighbourhood of the point $t = 0$ on the axis (recall that affine substitutions of argument preserve convexity). Since f is twice differentiable in a neighbourhood of x , g is twice differentiable in a neighbourhood of $t = 0$, so that $g''(0) = h^T f''(x)h \geq 0$ by Proposition C.2.1. \square

Now let us prove the "if" part, so that we are given that $h^T f''(x)h \geq 0$ for every $x \in \text{int } Q$ and every $h \in \mathbf{R}^n$, and we should prove that f is convex.

Let us first prove that f is convex on the interior Q' of the domain Q . By Theorem B.1.1, Q' is a convex set. Since, as it was already explained, the convexity of a function on a convex set is one-dimensional fact, all we should prove is that every one-dimensional function

$$g(t) = f(x + t(y - x)), \quad 0 \leq t \leq 1$$

(x and y are from Q') is convex on the segment $0 \leq t \leq 1$. Since f is continuous on $Q \supset Q'$, g is continuous on the segment; and since f is twice continuously differentiable on Q' , g is continuously differentiable on $(0, 1)$ with the second derivative

$$g''(t) = (y - x)^T f''(x + t(y - x))(y - x) \geq 0.$$

Consequently, g is convex on $[0, 1]$ (Propositions C.2.1.(ii) and C.2.2). Thus, f is convex on Q' . It remains to note that f , being convex on Q' and continuous on Q , is convex on Q by Proposition C.2.2. ■

Applying the combination rules preserving convexity to simple functions which pass the "infinitesimal" convexity tests, we can prove convexity of many complicated functions. Consider, e.g., an *exponential posynomial* – a function

$$f(x) = \sum_{i=1}^N c_i \exp\{a_i^T x\}$$

with positive coefficients c_i (this is why the function is called *posynomial*). How could we prove that the function is convex? This is immediate:

$\exp\{t\}$ is convex (since its second order derivative is positive and therefore the first derivative is monotone, as required by the infinitesimal convexity test for smooth functions of one variable);

consequently, all functions $\exp\{a_i^T x\}$ are convex (stability of convexity under affine substitutions of argument);

consequently, f is convex (stability of convexity under taking linear combinations with non-negative coefficients).

And if we were supposed to prove that the maximum of three posynomials is convex? Ok, we could add to our three steps the fourth, which refers to stability of convexity under taking pointwise supremum.

C.3 Gradient inequality

An extremely important property of a convex function is given by the following

Proposition C.3.1 [Gradient inequality] *Let f be a function taking finite values and the value $+\infty$, x be an interior point of the domain of f and Q be a convex set containing x . Assume that*

- f is convex on Q

and

- f is differentiable at x ,

and let $\nabla f(x)$ be the gradient of the function at x . Then the following inequality holds:

$$(\forall y \in Q) : \quad f(y) \geq f(x) + (y - x)^T \nabla f(x). \quad (\text{C.3.1})$$

Geometrically: the graph

$$\{(y, t) \in \mathbf{R}^{n+1} : y \in \text{Dom } f \cap Q, t = f(y)\}$$

of the function f restricted onto the set Q is above the graph

$$\{(y, t) \in \mathbf{R}^{n+1} : t = f(x) + (y - x)^T \nabla f(x)\}$$

of the linear form tangent to f at x .

Proof. Let $y \in Q$. There is nothing to prove if $y \notin \text{Dom } f$ (since there the right hand side in the gradient inequality is $+\infty$), same as there is nothing to prove when $y = x$. Thus, we can assume that $y \neq x$ and $y \in \text{Dom } f$. Let us set

$$y_\tau = x + \tau(y - x), \quad 0 < \tau \leq 1,$$

so that $y_1 = y$ and y_τ is an interior point of the segment $[x, y]$ for $0 < \tau < 1$. Now let us use the following extremely simple

Lemma C.3.1 *Let x, x', x'' be three distinct points with $x' \in [x, x'']$, and let f be convex and finite on $[x, x'']$. Then*

$$\frac{f(x') - f(x)}{\|x' - x\|_2} \leq \frac{f(x'') - f(x)}{\|x'' - x\|_2}. \quad (\text{C.3.2})$$

Proof of the Lemma. We clearly have

$$x' = x + \lambda(x'' - x), \quad \lambda = \frac{\|x' - x\|_2}{\|x'' - x\|_2} \in (0, 1)$$

or, which is the same,

$$x' = (1 - \lambda)x + \lambda x''.$$

From the convexity inequality

$$f(x') \leq (1 - \lambda)f(x) + \lambda f(x''),$$

or, which is the same,

$$f(x') - f(x) \leq \lambda(f(x'') - f(x')).$$

Dividing by λ and substituting the value of λ , we come to (C.3.2). \square

Applying the Lemma to the triple $x, x' = y_\tau, x'' = y$, we get

$$\frac{f(x + \tau(y - x)) - f(x)}{\tau \|y - x\|_2} \leq \frac{f(y) - f(x)}{\|y - x\|_2};$$

as $\tau \rightarrow +0$, the left hand side in this inequality, by the definition of the gradient, tends to $\|y - x\|_2^{-1}(y - x)^T \nabla f(x)$, and we get

$$\|y - x\|_2^{-1}(y - x)^T \nabla f(x) \leq \|y - x\|_2^{-1}(f(y) - f(x)),$$

or, which is the same,

$$(y - x)^T \nabla f(x) \leq f(y) - f(x);$$

this is exactly the inequality (C.3.1). ■

It is worthy of mentioning that in the case when Q is convex set with a nonempty interior and f is continuous on Q and differentiable on $\text{int } Q$, f is convex on Q if and only if the Gradient inequality (C.3.1) is valid for every pair $x \in \text{int } Q$ and $y \in Q$.

Indeed, the "only if" part, i.e., the implication

$$\text{convexity of } f \Rightarrow \text{Gradient inequality for all } x \in \text{int } Q \text{ and all } y \in Q$$

is given by Proposition C.3.1. To prove the "if" part, i.e., to establish the implication inverse to the above, assume that f satisfies the Gradient inequality for all $x \in \text{int } Q$ and all $y \in Q$, and let us verify that f is convex on Q . It suffices to prove that f is convex on the interior Q' of the set Q (see Proposition C.2.2; recall that by assumption f is continuous on Q and Q is convex). To prove that f is convex on Q' , note that Q' is convex (Theorem B.1.1) and that, due to the Gradient inequality, on Q' f is the upper bound of the family of affine (and therefore convex) functions:

$$f(y) = \sup_{x \in Q'} f_x(y), \quad f_x(y) = f(x) + (y - x)^T \nabla f(x).$$

■

C.4 Boundedness and Lipschitz continuity of a convex function

Convex functions possess nice local properties.

Theorem C.4.1 [local boundedness and Lipschitz continuity of convex function]

Let f be a convex function and let K be a closed and bounded set contained in the relative interior of the domain $\text{Dom } f$ of f . Then f is Lipschitz continuous on K – there exists constant L – the Lipschitz constant of f on K – such that

$$|f(x) - f(y)| \leq L\|x - y\|_2 \quad \forall x, y \in K. \quad (\text{C.4.1})$$

In particular, f is bounded on K .

Remark C.4.1 All three assumptions on K – (1) closedness, (2) boundedness, and (3) $K \subset \text{ri Dom } f$ – are essential, as it is seen from the following three examples:

- $f(x) = 1/x$, $\text{Dom } F = (0, +\infty)$, $K = (0, 1]$. We have (2), (3) but not (1); f is neither bounded, nor Lipschitz continuous on K .
- $f(x) = x^2$, $\text{Dom } f = \mathbf{R}$, $K = \mathbf{R}$. We have (1), (3) and not (2); f is neither bounded nor Lipschitz continuous on K .
- $f(x) = -\sqrt{x}$, $\text{Dom } f = [0, +\infty)$, $K = [0, 1]$. We have (1), (2) and not (3); f is not Lipschitz continuous on K ¹⁾, although is bounded. With properly chosen convex function f of two variables and non-polyhedral compact domain (e.g., with $\text{Dom } f$ being the unit circle), we could demonstrate also that lack of (3), even in presence of (1) and (2), may cause unboundedness of f at K as well.

Remark C.4.2 Theorem C.4.1 says that a convex function f is bounded on every compact (i.e., closed and bounded) subset of the relative interior of $\text{Dom } f$. In fact there is much stronger statement on the below boundedness of f : f is below bounded on any bounded subset of \mathbf{R}^n !.

Proof of Theorem C.4.1. We shall start with the following local version of the Theorem.

Proposition C.4.1 Let f be a convex function, and let \bar{x} be a point from the relative interior of the domain $\text{Dom } f$ of f . Then

(i) f is bounded at \bar{x} : there exists a positive r such that f is bounded in the r -neighbourhood $U_r(\bar{x})$ of \bar{x} in the affine span of $\text{Dom } f$:

$$\exists r > 0, C : |f(x)| \leq C \quad \forall x \in U_r(\bar{x}) = \{x \in \text{Aff}(\text{Dom } f) : \|x - \bar{x}\|_2 \leq r\};$$

(ii) f is Lipschitz continuous at \bar{x} , i.e., there exists a positive ρ and a constant L such that

$$|f(x) - f(x')| \leq L\|x - x'\|_2 \quad \forall x, x' \in U_\rho(\bar{x}).$$

Implication “Proposition C.4.1 \Rightarrow Theorem C.4.1” is given by standard Analysis reasoning. All we need is to prove that if K is a bounded and closed (i.e., a compact) subset of $\text{ri Dom } f$, then f is Lipschitz continuous on K (the boundedness of f on K is an evident consequence of its Lipschitz continuity on K and boundedness of K). Assume, on contrary, that f is not Lipschitz continuous on K ; then for every integer i there exists a pair of points $x_i, y_i \in K$ such that

$$f(x_i) - f(y_i) \geq i\|x_i - y_i\|_2. \tag{C.4.2}$$

Since K is compact, passing to a subsequence we can ensure that $x_i \rightarrow x \in K$ and $y_i \rightarrow y \in K$. By Proposition C.4.1 the case $x = y$ is impossible – by Proposition f is Lipschitz continuous in a neighbourhood B of $x = y$; since $x_i \rightarrow x, y_i \rightarrow y$, this neighbourhood should contain all x_i and y_i with large enough indices i ; but then, from the Lipschitz continuity of f in B , the ratios $(f(x_i) - f(y_i))/\|x_i - y_i\|_2$ form a bounded sequence, which we know is not the case. Thus, the case $x = y$ is impossible. The case $x \neq y$ is “even less possible” – since, by Proposition, f is continuous on $\text{Dom } f$ at both the points x and y (note that Lipschitz continuity at a point clearly implies the usual continuity at it), so that we would have $f(x_i) \rightarrow f(x)$ and $f(y_i) \rightarrow f(y)$

¹⁾indeed, we have $\lim_{t \rightarrow +0} \frac{f(0) - f(t)}{t} = \lim_{t \rightarrow +0} t^{-1/2} = +\infty$, while for a Lipschitz continuous f the ratios $t^{-1}(f(0) - f(t))$ should be bounded

as $i \rightarrow \infty$. Thus, the left hand side in (C.4.2) remains bounded as $i \rightarrow \infty$. In the right hand side one factor $-i$ tends to ∞ , and the other one has a nonzero limit $\|x - y\|$, so that the right hand side tends to ∞ as $i \rightarrow \infty$; this is the desired contradiction. \square

Proof of Proposition C.4.1.

1⁰. We start with proving the *above boundedness* of f in a neighbourhood of \bar{x} . This is immediate: we know that there exists a neighbourhood $U_{\bar{r}}(\bar{x})$ which is contained in $\text{Dom } f$ (since, by assumption, \bar{x} is a relative interior point of $\text{Dom } f$). Now, we can find a small simplex Δ of the dimension $m = \dim \text{Aff}(\text{Dom } f)$ with the vertices x_0, \dots, x_m in $U_{\bar{r}}(\bar{x})$ in such a way that \bar{x} will be a convex combination of the vectors x_i with positive coefficients, even with the coefficients $1/(m+1)$:

$$\bar{x} = \sum_{i=0}^m \frac{1}{m+1} x_i \quad ^2).$$

We know that \bar{x} is the point from the relative interior of Δ (Exercise B.8); since Δ spans the same affine subspace as $\text{Dom } f$, it means that Δ contains $U_r(\bar{x})$ with certain $r > 0$. Now, we have

$$\Delta = \left\{ \sum_{i=0}^m \lambda_i x_i : \lambda_i \geq 0, \sum_i \lambda_i = 1 \right\}$$

so that in Δ f is bounded from above by the quantity $\max_{0 \leq i \leq m} f(x_i)$ by Jensen's inequality:

$$f\left(\sum_{i=0}^m \lambda_i x_i\right) \leq \sum_{i=0}^m \lambda_i f(x_i) \leq \max_i f(x_i).$$

Consequently, f is bounded from above, by the same quantity, in $U_r(\bar{x})$.

2⁰. Now let us prove that if f is above bounded, by some C , in $U_r = U_r(\bar{x})$, then it in fact is below bounded in this neighbourhood (and, consequently, is bounded in U_r). Indeed, let $x \in U_r$, so that $x \in \text{Aff}(\text{Dom } f)$ and $\|x - \bar{x}\|_2 \leq r$. Setting $x' = \bar{x} - [x - \bar{x}] = 2\bar{x} - x$, we get $x' \in \text{Aff}(\text{Dom } f)$ and $\|x' - \bar{x}\|_2 = \|x - \bar{x}\|_2 \leq r$, so that $x' \in U_r$. Since $\bar{x} = \frac{1}{2}[x + x']$, we have

$$2f(\bar{x}) \leq f(x) + f(x'),$$

whence

$$f(x) \geq 2f(\bar{x}) - f(x') \geq 2f(\bar{x}) - C, \quad x \in U_r(\bar{x}),$$

and f is indeed below bounded in U_r .

(i) is proved.

3⁰. (ii) is an immediate consequence of (i) and Lemma C.3.1. Indeed, let us prove that f is Lipschitz continuous in the neighbourhood $U_{r/2}(\bar{x})$, where $r > 0$ is such that f is bounded in $U_r(\bar{x})$ (we already know from (i) that the required r does exist). Let $|f| \leq C$ in U_r , and let

²to see that the required Δ exists, let us act as follows: first, the case of $\text{Dom } f$ being a singleton is evident, so that we can assume that $\text{Dom } f$ is a convex set of dimension $m \geq 1$. Without loss of generality, we may assume that $\bar{x} = 0$, so that $0 \in \text{Dom } f$ and therefore $\text{Aff}(\text{Dom } f) = \text{Lin}(\text{Dom } f)$. By Linear Algebra, we can find m vectors y_1, \dots, y_m in $\text{Dom } f$ which form a basis in $\text{Lin}(\text{Dom } f) = \text{Aff}(\text{Dom } f)$. Setting $y_0 = -\sum_{i=1}^m y_i$ and taking into account that $0 = \bar{x} \in \text{ri Dom } f$, we can find $\epsilon > 0$ such that the vectors $x_i = \epsilon y_i$, $i = 0, \dots, m$, belong to $U_{\bar{r}}(\bar{x})$. By construction, $\bar{x} = 0 = \frac{1}{m+1} \sum_{i=0}^m x_i$.

$x, x' \in U_{r/2}$, $x \neq x'$. Let us extend the segment $[x, x']$ through the point x' until it reaches, at certain point x'' , the (relative) boundary of U_r . We have

$$x' \in (x, x''); \quad \|x'' - \bar{x}\|_2 = r.$$

From (C.3.2) we have

$$f(x') - f(x) \leq \|x' - x\|_2 \frac{f(x'') - f(x)}{\|x'' - x\|_2}.$$

The second factor in the right hand side does not exceed the quantity $(2C)/(r/2) = 4C/r$; indeed, the numerator is, in absolute value, at most $2C$ (since $|f|$ is bounded by C in U_r and both x, x'' belong to U_r), and the denominator is at least $r/2$ (indeed, x is at the distance at most $r/2$ from \bar{x} , and x'' is at the distance exactly r from \bar{x} , so that the distance between x and x'' , by the triangle inequality, is at least $r/2$). Thus, we have

$$f(x') - f(x) \leq (4C/r)\|x' - x\|_2, \quad x, x' \in U_{r/2};$$

swapping x and x' , we come to

$$f(x) - f(x') \leq (4C/r)\|x' - x\|_2,$$

whence

$$|f(x) - f(x')| \leq (4C/r)\|x - x'\|_2, \quad x, x' \in U_{r/2},$$

as required in (ii). ■

C.5 Maxima and minima of convex functions

As it was already mentioned, optimization problems involving convex functions possess nice theoretical properties. One of the most important of these properties is given by the following

Theorem C.5.1 [“Unimodality”] *Let f be a convex function on a convex set $Q \subset \mathbf{R}^n$, and let $x^* \in Q \cap \text{Dom } f$ be a local minimizer of f on Q :*

$$(\exists r > 0) : \quad f(y) \geq f(x^*) \quad \forall y \in Q, \|y - x\|_2 < r. \quad (\text{C.5.1})$$

Then x^ is a global minimizer of f on Q :*

$$f(y) \geq f(x^*) \quad \forall y \in Q. \quad (\text{C.5.2})$$

Moreover, the set $\underset{Q}{\text{Argmin}} f$ of all local (\equiv global) minimizers of f on Q is convex.

If f is strictly convex (i.e., the convexity inequality $f(\lambda x + (1 - \lambda)y) \leq \lambda f(x) + (1 - \lambda)f(y)$ is strict whenever $x \neq y$ and $\lambda \in (0, 1)$), then the above set is either empty or is a singleton.

Proof. 1) Let x^* be a local minimizer of f on Q and $y \in Q$, $y \neq x^*$; we should prove that $f(y) \geq f(x^*)$. There is nothing to prove if $f(y) = +\infty$, so that we may assume that $y \in \text{Dom } f$. Note that also $x^* \in \text{Dom } f$ for sure – by definition of a local minimizer.

For all $\tau \in (0, 1)$ we have, by Lemma C.3.1,

$$\frac{f(x^* + \tau(y - x^*)) - f(x^*)}{\tau\|y - x^*\|_2} \leq \frac{f(y) - f(x^*)}{\|y - x^*\|_2}.$$

Since x^* is a local minimizer of f , the left hand side in this inequality is nonnegative for all small enough values of $\tau > 0$. We conclude that the right hand side is nonnegative, i.e., $f(y) \geq f(x^*)$.

□

2) To prove convexity of $\underset{Q}{\operatorname{Argmin}} f$, note that $\underset{Q}{\operatorname{Argmin}} f$ is nothing but the level set $\operatorname{lev}_\alpha(f)$ of f associated with the minimal value $\min_Q f$ of f on Q ; as a level set of a convex function, this set is convex (Proposition C.1.4).

3) To prove that the set $\underset{Q}{\operatorname{Argmin}} f$ associated with a strictly convex f is, if nonempty, a singleton, note that if there were two distinct minimizers x', x'' , then, from strict convexity, we would have

$$f\left(\frac{1}{2}x' + \frac{1}{2}x''\right) < \frac{1}{2}[f(x') + f(x'')] == \min_Q f,$$

which clearly is impossible - the argument in the left hand side is a point from Q ! ■

Another pleasant fact is that in the case of differentiable convex functions the known from Calculus necessary optimality condition (the Fermat rule) is sufficient for global optimality:

Theorem C.5.2 [Necessary and sufficient optimality condition for a differentiable convex function]

Let f be convex function on convex set $Q \subset \mathbf{R}^n$, and let x^* be an interior point of Q . Assume that f is differentiable at x^* . Then x^* is a minimizer of f on Q if and only if

$$\nabla f(x^*) = 0.$$

Proof. As a necessary condition for local optimality, the relation $\nabla f(x^*) = 0$ is known from Calculus; it has nothing in common with convexity. The essence of the matter is, of course, the sufficiency of the condition $\nabla f(x^*) = 0$ for global optimality of x^* in the case of convex f . This sufficiency is readily given by the Gradient inequality (C.3.1): by virtue of this inequality and due to $\nabla f(x^*) = 0$,

$$f(y) \geq f(x^*) + (y - x^*)^\top \nabla f(x^*) = f(x^*)$$

for all $y \in Q$. ■

A natural question is what happens if x^* in the above statement is not necessarily an interior point of Q . Thus, assume that x^* is an arbitrary point of a convex set Q and that f is convex on Q and differentiable at x^* (the latter means exactly that $\operatorname{Dom} f$ contains a neighbourhood of x^* and f possesses the first order derivative at x^*). Under these assumptions, when x^* is a minimizer of f on Q ?

The answer is as follows: let

$$T_Q(x^*) = \{h \in \mathbf{R}^n : x^* + th \in Q \quad \forall \text{ small enough } t > 0\}$$

be the *radial cone* of Q at x^* ; geometrically, this is the set of all directions leading from x^* inside Q , so that a small enough positive step from x^* along the direction keeps the point in Q . From the convexity of Q it immediately follows that the radial cone indeed is a convex cone (not necessary closed). E.g., when x^* is an interior point of Q , then the radial cone to Q at x^* clearly is the entire \mathbf{R}^n . A more interesting example is the radial cone to a polyhedral set

$$Q = \{x : a_i^T x \leq b_i, i = 1, \dots, m\}; \tag{C.5.3}$$

for $x^* \in Q$ the corresponding radial cone clearly is the polyhedral cone

$$\{h : a_i^T h \leq 0 \quad \forall i : a_i^T x^* = b_i\} \tag{C.5.4}$$

corresponding to the active at x^* (i.e., satisfied at the point as equalities rather than as strict inequalities) constraints $a_i^T x \leq b_i$ from the description of Q .

Now, for the functions in question the necessary and sufficient condition for x^* to be a minimizer of f on Q is as follows:

Proposition C.5.1 *Let Q be a convex set, let $x^* \in Q$, and let f be a convex on Q function which is differentiable at x^* . The necessary and sufficient condition for x^* to be a minimizer of f on Q is that the derivative of f taken at x^* along every direction from $T_Q(x^*)$ should be nonnegative:*

$$h^T \nabla f(x^*) \geq 0 \quad \forall h \in T_Q(x^*).$$

Proof is immediate. The necessity is an evident fact which has nothing in common with convexity: assuming that x^* is a local minimizer of f on Q , we note that if there were $h \in T_Q(x^*)$ with $h^T \nabla f(x^*) < 0$, then we would have

$$f(x^* + th) < f(x^*)$$

for all small enough positive t . On the other hand, $x^* + th \in Q$ for all small enough positive t due to $h \in T_Q(x^*)$. Combining these observations, we conclude that in every neighbourhood of x^* there are points from Q with strictly better than the one at x^* values of f ; this contradicts the assumption that x^* is a local minimizer of f on Q .

The sufficiency is given by the Gradient Inequality, exactly as in the case when x^* is an interior point of Q . ■

Proposition C.5.1 says that whenever f is convex on Q and differentiable at $x^* \in Q$, the necessary and sufficient condition for x^* to be a minimizer of f on Q is that the linear form given by the gradient $\nabla f(x^*)$ of f at x^* should be nonnegative at all directions from the radial cone $T_Q(x^*)$. The linear forms nonnegative at all directions from the radial cone also form a cone; it is called the cone *normal* to Q at x^* and is denoted $N_Q(x^*)$. Thus, Proposition says that the necessary and sufficient condition for x^* to minimize f on Q is the inclusion $\nabla f(x^*) \in N_Q(x^*)$. What does this condition actually mean, it depends on what is the normal cone: whenever we have an explicit description of it, we have an explicit form of the optimality condition.

E.g., when $T_Q(x^*) = \mathbf{R}^n$ (it is the same as to say that x^* is an interior point of Q), then the normal cone is comprised of the linear forms nonnegative at the entire space, i.e., it is the trivial cone $\{0\}$; consequently, for the case in question the optimality condition becomes the Fermat rule $\nabla f(x^*) = 0$, as we already know.

When Q is the polyhedral set (C.5.3), the normal cone is the polyhedral cone (C.5.4); it is comprised of all directions which have nonpositive inner products with all a_i coming from the active, in the aforementioned sense, constraints. The normal cone is comprised of all vectors which have nonnegative inner products with all these directions, i.e., of vectors a such that the inequality $h^T a \geq 0$ is a consequence of the inequalities $h^T a_i \leq 0$, $i \in I(x^*) \equiv \{i : a_i^T x^* = b_i\}$. From the Homogeneous Farkas Lemma we conclude that the normal cone is simply the conic hull of the vectors $-a_i$, $i \in I(x^*)$. Thus, in the case in question (*) reads:

$x^* \in Q$ is a minimizer of f on Q if and only if there exist nonnegative reals λ_i^* associated with “active” (those from $I(x^*)$) values of i such that

$$\nabla f(x^*) + \sum_{i \in I(x^*)} \lambda_i^* a_i = 0.$$

These are the famous *Karush-Kuhn-Tucker* optimality conditions; these conditions are necessary for optimality in an essentially wider situation.

The indicated results demonstrate that the fact that a point $x^* \in \text{Dom } f$ is a global minimizer of a convex function f depends only on the local behaviour of f at x^* . This is not the case with maximizers of a convex function. First of all, such a maximizer, if exists, in all nontrivial cases should belong to the boundary of the domain of the function:

Theorem C.5.3 *Let f be convex, and let Q be the domain of f . Assume that f attains its maximum on Q at a point x^* from the relative interior of Q . Then f is constant on Q .*

Proof. Let $y \in Q$; we should prove that $f(y) = f(x^*)$. There is nothing to prove if $y = x^*$, so that we may assume that $y \neq x^*$. Since, by assumption, $x^* \in \text{ri } Q$, we can extend the segment $[x^*, y]$ through the endpoint x^* , keeping the left endpoint of the segment in Q ; in other words, there exists a point $y' \in Q$ such that x^* is an interior point of the segment $[y', y]$:

$$x^* = \lambda y' + (1 - \lambda)y$$

for certain $\lambda \in (0, 1)$. From the definition of convexity

$$f(x^*) \leq \lambda f(y') + (1 - \lambda)f(y).$$

Since both $f(y')$ and $f(y)$ do not exceed $f(x^*)$ (x^* is a maximizer of f on Q !) and both the weights λ and $1 - \lambda$ are strictly positive, the indicated inequality can be valid only if $f(y') = f(y) = f(x^*)$. ■

The next theorem gives further information on maxima of convex functions:

Theorem C.5.4 *Let f be a convex function on \mathbf{R}^n and E be a subset of \mathbf{R}^n . Then*

$$\sup_{\text{Conv}E} f = \sup_E f. \quad (\text{C.5.5})$$

In particular, if $S \subset \mathbf{R}^n$ is convex and compact set, then the supremum of f on S is equal to the supremum of f on the set of extreme points of S :

$$\sup_S f = \sup_{\text{Ext}(S)} f \quad (\text{C.5.6})$$

Proof. To prove (C.5.5), let $x \in \text{Conv}E$, so that x is a convex combination of points from E (Theorem B.1.4 on the structure of convex hull):

$$x = \sum_i \lambda_i x_i \quad [x_i \in E, \lambda_i \geq 0, \sum_i \lambda_i = 1].$$

Applying Jensen's inequality (Proposition C.1.3), we get

$$f(x) \leq \sum_i \lambda_i f(x_i) \leq \sum_i \lambda_i \sup_E f = \sup_E f,$$

so that the left hand side in (C.5.5) is \leq the right hand one; the inverse inequality is evident, since $\text{Conv}E \supset E$. □

To derive (C.5.6) from (C.5.5), it suffices to note that from the Krein-Milman Theorem (Theorem B.2.10) for a convex compact set S one has $S = \text{ConvExt}(S)$. ■

The last theorem on maxima of convex functions is as follows:

Theorem C.5.5 *Let f be a convex function such that the domain Q of f is closed and does not contain lines. Then*

(i) *If the set*

$$\underset{Q}{\text{Argmax}} f \equiv \{x \in Q : f(x) \geq f(y) \forall y \in Q\}$$

of global maximizers of f is nonempty, then it intersects the set $\text{Ext}(Q)$ of the extreme points of Q , so that at least one of the maximizers of f is an extreme point of Q ;

(ii) *If the set Q is polyhedral and f is above bounded on Q , then the maximum of f on Q is achieved: $\underset{Q}{\text{Argmax}} f \neq \emptyset$.*

Proof. Let us start with (i). We shall prove this statement by induction on the dimension of Q . The base $\dim Q = 0$, i.e., the case of a singleton Q , is trivial, since here $Q = \text{Ext}Q = \underset{Q}{\text{Argmax}} f$. Now assume that the statement is valid for the case of $\dim Q \leq p$, and let us prove that it is valid also for the case of $\dim Q = p + 1$. Let us first verify that the set $\underset{Q}{\text{Argmax}} f$ intersects with the (relative) boundary of Q . Indeed, let $x \in \underset{Q}{\text{Argmax}} f$. There is nothing to prove if x itself is a relative boundary point of Q ; and if x is not a boundary point, then, by Theorem C.5.3, f is constant on Q , so that $\underset{Q}{\text{Argmax}} f = Q$; and since Q is closed, every relative boundary point of Q (such a point does exist, since Q does not contain lines and is of positive dimension) is a maximizer of f on Q , so that here again $\underset{Q}{\text{Argmax}} f$ intersects $\partial_{\text{ri}} Q$.

Thus, among the maximizers of f there exists at least one, let it be x , which belongs to the relative boundary of Q . Let H be the hyperplane which supports Q at x (see Section B.2.6), and let $Q' = Q \cap H$. The set Q' is closed and convex (since Q and H are), nonempty (it contains x) and does not contain lines (since Q does not). We have $\max_Q f = f(x) = \max_{Q'} f$ (note that $Q' \subset Q$), whence

$$\emptyset \neq \underset{Q'}{\text{Argmax}} f \subset \underset{Q}{\text{Argmax}} f.$$

Same as in the proof of the Krein-Milman Theorem (Theorem B.2.10), we have $\dim Q' < \dim Q$. In view of this inequality we can apply to f and Q' our inductive hypothesis to get

$$\text{Ext}(Q') \cap \underset{Q'}{\text{Argmax}} f \neq \emptyset.$$

Since $\text{Ext}(Q') \subset \text{Ext}(Q)$ by Lemma B.2.4 and, as we just have seen, $\underset{Q'}{\text{Argmax}} f \subset \underset{Q}{\text{Argmax}} f$, we conclude that the set $\underset{Q}{\text{Ext}}(Q) \cap \underset{Q}{\text{Argmax}} f$ is not smaller than $\text{Ext}(Q') \cap \underset{Q'}{\text{Argmax}} f$ and is therefore nonempty, as required. \square

To prove (ii), let us use the known to us from Lecture 4 results on the structure of a polyhedral convex set:

$$Q = \text{Conv}(V) + \text{Cone}(R),$$

where V and R are finite sets. We are about to prove that the upper bound of f on Q is exactly the maximum of f on the finite set V :

$$\forall x \in Q : f(x) \leq \max_{v \in V} f(v). \quad (\text{C.5.7})$$

This will mean, in particular, that f attains its maximum on Q – e.g., at the point of V where f attains its maximum on V .

To prove the announced statement, I first claim that if f is above bounded on Q , then every direction $r \in \text{Cone}(R)$ is *descent* for f , i.e., is such that every step in this direction taken from every point $x \in Q$ decreases f :

$$f(x + tr) \leq f(x) \quad \forall x \in Q \forall t \geq 0. \quad (\text{C.5.8})$$

Indeed, if, on contrary, there were $x \in Q$, $r \in R$ and $t \geq 0$ such that $f(x + tr) > f(x)$, we would have $t > 0$ and, by Lemma C.3.1,

$$f(x + sr) \geq f(x) + \frac{s}{t}(f(x + tr) - f(x)), \quad s \geq t.$$

Since $x \in Q$ and $r \in \text{Cone}(R)$, $x + sr \in Q$ for all $s \geq 0$, and since f is above bounded on Q , the left hand side in the latter inequality is above bounded, while the right hand one, due to $f(x + tr) > f(x)$, goes to $+\infty$ as $s \rightarrow \infty$, which is the desired contradiction.

Now we are done: to prove (C.5.7), note that a generic point $x \in Q$ can be represented as

$$x = \sum_{v \in V} \lambda_v v + r \quad [r \in \text{Cone}(R); \sum_v \lambda_v = 1, \lambda_v \geq 0],$$

and we have

$$\begin{aligned} f(x) &= f\left(\sum_{v \in V} \lambda_v v + r\right) \\ &\leq f\left(\sum_{v \in V} \lambda_v v\right) \quad [\text{by (C.5.8)}] \\ &\leq \sum_{v \in V} \lambda_v f(v) \quad [\text{Jensen's Inequality}] \\ &\leq \max_{v \in V} f(v) \quad \blacksquare \end{aligned}$$

C.6 Subgradients and Legendre transformation

C.6.1 Proper functions and their representation

According to one of two equivalent definitions, a convex function f on \mathbf{R}^n is a function taking values in $\mathbf{R} \cup \{+\infty\}$ such that the epigraph

$$\text{Epi}(f) = \{(t, x) \in \mathbf{R}^{n+1} : t \geq f(x)\}$$

is a nonempty convex set. Thus, there is no essential difference between convex functions and convex sets: convex function generates a convex set – its epigraph – which of course remembers everything about the function. And the only specific property of the epigraph as a convex set is that it has a recessive direction – namely, $e = (1, 0)$ – such that the intersection of the epigraph with every line directed by h is either empty, or is a closed ray. Whenever a nonempty convex set possesses such a property with respect to certain direction, it can be represented, in properly chosen coordinates, as the epigraph of some convex function. Thus, a convex function is, basically, nothing but a way to look, in the literal meaning of the latter verb, at a convex set.

Now, we know that “actually good” convex sets are closed ones: they possess a lot of important properties (e.g., admit a good outer description) which are not shared by arbitrary convex sets. It means that among convex functions there also are “actually good” ones – those with closed epigraphs. Closedness of the epigraph can be “translated” to the functional language and there becomes a special kind of continuity – *lower semicontinuity*:

Definition C.6.1 [Lower semicontinuity] Let f be a function (not necessarily convex) defined on \mathbf{R}^n and taking values in $\mathbf{R} \cup \{+\infty\}$. We say that f is lower semicontinuous at a point \bar{x} , if for every sequence of points $\{x_i\}$ converging to \bar{x} one has

$$f(\bar{x}) \leq \liminf_{i \rightarrow \infty} f(x_i)$$

(here, of course, \liminf of a sequence with all terms equal to $+\infty$ is $+\infty$).

f is called lower semicontinuous, if it is lower semicontinuous at every point.

A trivial example of a lower semicontinuous function is a continuous one. Note, however, that a semicontinuous function is not obliged to be continuous; what it is obliged, is to make only “jumps down”. E.g., the function

$$f(x) = \begin{cases} 0, & x \neq 0 \\ a, & x = 0 \end{cases}$$

is lower semicontinuous if $a \leq 0$ (“jump down at $x = 0$ or no jump at all”), and is not lower semicontinuous if $a > 0$ (“jump up”).

The following statement links lower semicontinuity with the geometry of the epigraph:

Proposition C.6.1 A function f defined on \mathbf{R}^n and taking values from $\mathbf{R} \cup \{+\infty\}$ is lower semicontinuous if and only if its epigraph is closed (e.g., due to its emptiness).

I shall not prove this statement, same as most of other statements in this Section; the reader definitely is able to restore (very simple) proofs I am skipping.

An immediate consequence of the latter proposition is as follows:

Corollary C.6.1 *The upper bound*

$$f(x) = \sup_{\alpha \in \mathcal{A}} f_\alpha(x)$$

of arbitrary family of lower semicontinuous functions is lower semicontinuous.

[from now till the end of the Section, if the opposite is not explicitly stated, “a function” means “a function defined on the entire \mathbf{R}^n and taking values in $\mathbf{R} \cup \{+\infty\}$ ”]

Indeed, the epigraph of the upper bound is the intersection of the epigraphs of the functions forming the bound, and the intersection of closed sets always is closed.

Now let us look at convex lower semicontinuous functions; according to our general convention, “convex” means “satisfying the convexity inequality and finite at least at one point”, or, which is the same, “with convex nonempty epigraph”; and as we just have seen, “lower semicontinuous” means “with closed epigraph”. Thus, we are interested in functions with closed convex nonempty epigraphs; to save words, let us call these functions proper.

What we are about to do is to translate to the functional language several constructions and results related to convex sets. In the usual life, a translation (e.g. of poetry) typically results in something less rich than the original; in contrast to this, in mathematics this is a powerful source of new ideas and constructions.

“Outer description” of a proper function. We know that a closed convex set is intersection of closed half-spaces. What does this fact imply when the set is the epigraph of a proper function f ? First of all, note that the epigraph is not a completely arbitrary convex set: it has a recessive direction $e = (1, 0)$ – the basic orth of the t -axis in the space of variables $t \in \mathbf{R}$, $x \in \mathbf{R}^n$ where the epigraph lives. This direction, of course, should be recessive for every closed half-space

$$(*) \quad \Pi = \{(t, x) : \alpha t \geq d^T x - a\} \quad [|\alpha| + |d| > 0]$$

containing $\text{Epi}(f)$ (note that what is written in the right hand side of the latter relation, is one of many universal forms of writing down a general nonstrict linear inequality in the space where the epigraph lives; this is the form the most convenient for us now). Thus, e should be a recessive direction of $\Pi \supset \text{Epi}(f)$; as it is immediately seen, recessivity of e for Π means exactly that $\alpha \geq 0$. Thus, speaking about closed half-spaces containing $\text{Epi}(f)$, we in fact are considering some of the half-spaces $(*)$ with $\alpha \geq 0$.

Now, there are two essentially different possibilities for α to be nonnegative – (A) to be positive, and (B) to be zero. In the case of (B) the boundary hyperplane of Π is “vertical” – it is parallel to e , and in fact it “bounds” only $x - \Pi$ is comprised of all pairs (t, x) with x belonging to certain half-space in the x -subspace and t being arbitrary real. These “vertical” subspaces will be of no interest for us.

The half-spaces which indeed are of interest for us are the “nonvertical” ones: those given by the case (A), i.e., with $\alpha > 0$. For a non-vertical half-space Π , we always can divide the inequality defining Π by α and to make $\alpha = 1$. Thus, a “nonvertical” candidate to the role of a closed half-space containing $\text{Epi}(f)$ always can be written down as

$$(**) \quad \Pi = \{(t, x) : t \geq d^T x - a\},$$

i.e., can be represented as the epigraph of an affine function of x .

Now, when such a candidate indeed is a half-space containing $\text{Epi}(f)$? The answer is clear: it is the case if and only if the affine function $d^T x - a$ everywhere in \mathbf{R}^n is $\leq f(\cdot)$ – as we shall say, “is an affine minorant of f ”; indeed, the smaller is the epigraph, the larger is the function. If we knew that $\text{Epi}(f)$ – which definitely is the intersection of all closed half-spaces containing $\text{Epi}(f)$ – is in fact the intersection of already nonvertical closed half-spaces containing $\text{Epi}(f)$, or, which is the same, the intersection of the epigraphs of all affine minorants of f , we would be able to get a nice and nontrivial result:

(!) a proper convex function is the upper bound of affine functions – all its affine minorants.

(indeed, we already know that it is the same – to say that a function is an upper bound of certain family of functions, and to say that the epigraph of the function is the intersection of the epigraphs of the functions of the family).

(!) indeed is true:

Proposition C.6.2 A proper convex function f is the upper bound of all its affine minorants. Moreover, at every point $\bar{x} \in \text{ri Dom } f$ from the relative interior of the domain of f f is even not the upper bound, but simply the maximum of its minorants: there exists an affine function $f_{\bar{x}}(x)$ which is $\leq f(x)$ everywhere in \mathbf{R}^n and is equal to f at $x = \bar{x}$.

Proof. I. We start with the “Moreover” part of the statement; this is the key to the entire statement. Thus, we are about to prove that if $\bar{x} \in \text{ri Dom } f$, then there exists an affine function $f_{\bar{x}}(x)$ which is everywhere $\leq f(x)$, and at $x = \bar{x}$ the inequality becomes an equality.

I.1⁰ First of all, we easily can reduce the situation to the one when $\text{Dom } f$ is full-dimensional. Indeed, by shifting f we may make the affine span $\text{Aff}(\text{Dom } f)$ of the domain of f to be a linear subspace L in \mathbf{R}^n ; restricting f onto this linear subspace, we clearly get a proper function on L . If we believe that our statement is true for the case when the domain of f is full-dimensional, we can conclude that there exists an affine function

$$d^T x - a \quad [x \in L]$$

on L ($d \in L$) such that

$$f(x) \geq d^T x - a \quad \forall x \in L; f(\bar{x}) = d^T \bar{x} - a.$$

The affine function we get clearly can be extended, by the same formula, from L on the entire \mathbf{R}^n and is a minorant of f on the entire \mathbf{R}^n – outside of $L \supset \text{Dom } f$ f simply is $+\infty$! This minorant on \mathbf{R}^n is exactly what we need.

I.2⁰. Now let us prove that our statement is valid when $\text{Dom } f$ is full-dimensional, so that \bar{x} is an interior point of the domain of f . Let us look at the point $y = (f(\bar{x}), \bar{x})$. This is a point from the epigraph of f , and I claim that it is a point from the relative boundary of the epigraph. Indeed, if y were a relative interior point of $\text{Epi}(f)$, then, taking $y' = y + e$, we would get a segment $[y', y]$ contained in $\text{Epi}(f)$; since the endpoint y of the segment is assumed to be relative interior for $\text{Epi}(f)$, we could extend this segment a little through this endpoint, not leaving $\text{Epi}(f)$; but this clearly is impossible, since the t -coordinate of the new endpoint would be $< f(\bar{x})$, and the x -component of it still would be \bar{x} .

Thus, y is a point from the relative boundary of $\text{Epi}(f)$. Now I claim that y' is an interior point of $\text{Epi}(f)$. This is immediate: we know from Theorem C.4.1 that f is continuous at \bar{x} , so that there exists a neighbourhood U of \bar{x} in $\text{Aff}(\text{Dom } f) = \mathbf{R}^n$ such that $f(x) \leq f(\bar{x} + 0.5)$ whenever $x \in U$, or, in other words, the set

$$V = \{(t, x) : x \in U, t > f(\bar{x}) + 0.5\}$$

is contained in $\text{Epi}(f)$; but this set clearly contains a neighbourhood of y' in \mathbf{R}^{n+1} .

Now let us look at the supporting linear form to $\text{Epi}(f)$ at the point y of the relative boundary of $\text{Epi}(f)$. This form gives us a linear inequality on \mathbf{R}^{n+1} which is satisfied everywhere on $\text{Epi}(f)$ and becomes equality at y ; besides this, the inequality is not equality identically on $\text{Epi}(f)$, it is strict somewhere on $\text{Epi}(f)$. Without loss of generality we may assume that the inequality is of the form

$$(+) \quad \alpha t \geq d^T x - a.$$

Now, since our inequality is satisfied at $y' = y + e$ and becomes equality at $(t, x) = y$, α should be ≥ 0 ; it cannot be 0, since in the latter case the inequality in question would be equality also at $y' \in \text{int Epi}(f)$. But a linear inequality which is satisfied at a convex set and is *equality* at an *interior* point of the set is trivial – coming from the zero linear form (this is exactly the statement that a linear form attaining its minimum on a convex set at a point from the relative interior of the set is constant on the set and on its affine hull).

Thus, inequality (+) which is satisfied on $\text{Epi}(f)$ and becomes equality at y is an inequality with $\alpha > 0$. Let us divide both sides of the inequality by α ; we get a new inequality of the form

$$(&) \quad t \geq d^T x - a$$

(I keep the same notation for the right hand side coefficients – we never will come back to the old coefficients); this inequality is valid on $\text{Epi}(f)$ and is equality at $y = (f(\bar{x}), \bar{x})$. Since the inequality is valid on $\text{Epi}(f)$, it is valid at every pair (t, x) with $x \in \text{Dom } f$ and $t = f(x)$:

$$(\#) \quad f(x) \geq d^T x - a \quad \forall x \in \text{Dom } f;$$

so that the right hand side is an affine minorant of f on $\text{Dom } f$ and therefore – on \mathbf{R}^n ($f = +\infty$ outside $\text{Dom } f$). It remains to note that (#) is equality at \bar{x} , since (&) is equality at y . \square

II. We have proved that if \mathcal{F} if the set of all affine functions which are minorants of f , then the function

$$\bar{f}(x) = \sup_{\phi \in \mathcal{F}} \phi(x)$$

is equal to f on $\text{ri Dom } f$ (and at x from the latter set in fact sup in the right hand side can be replaced with max); to complete the proof of the Proposition, we should prove that \bar{f} is equal to f also outside $\text{ri Dom } f$.

II.1⁰. Let us first prove that \bar{f} is equal to f outside $\text{cl Dom } f$, or. which is the same, prove that $\bar{f}(x) = +\infty$ outside $\text{cl Dom } f$. This is easy: is \bar{x} is a point outside $\text{cl Dom } f$, it can be strongly separated from $\text{Dom } f$, see Separation Theorem (ii) (Theorem B.2.9). Thus, there exists $z \in \mathbf{R}^n$ such that

$$z^T \bar{x} \geq z^T x + \zeta \quad \forall x \in \text{Dom } f \quad [\zeta > 0]. \quad (\text{C.6.1})$$

Besides this, we already know that there exists at least one affine minorant of f , or, which is the same, there exist a and d such that

$$f(x) \geq d^T x - a \quad \forall x \in \text{Dom } f. \quad (\text{C.6.2})$$

Let us add to (C.6.2) inequality (C.6.1) multiplied by positive weight λ ; we get

$$f(x) \geq \phi_\lambda(x) \equiv (d + \lambda z)^T x + [\lambda \zeta - a - \lambda z^T \bar{x}] \quad \forall x \in \text{Dom } f.$$

This inequality clearly says that $\phi_\lambda(\cdot)$ is an affine minorant of f on \mathbf{R}^n for every $\lambda > 0$. The value of this minorant at $x = \bar{x}$ is equal to $d^T \bar{x} - a + \lambda \zeta$ and therefore it goes to $+\infty$ as $\lambda \rightarrow +\infty$. We see that the upper bound of affine minorants of f at \bar{x} indeed is $+\infty$, as claimed.

II.2⁰. Thus, we know that the upper bound \bar{f} of all affine minorants of f is equal to f everywhere on the relative interior of $\text{Dom } f$ and everywhere outside the closure of $\text{Dom } f$; all we should prove that this equality is also valid at the points of the relative boundary of $\text{Dom } f$. Let \bar{x} be such a point. There is nothing to prove if $\bar{f}(\bar{x}) = +\infty$, since by construction \bar{f} is everywhere $\leq f$. Thus, we should prove that if $\bar{f}(\bar{x}) = c < \infty$, then $f(\bar{x}) = c$. Since $\bar{f} \leq f$ everywhere, to prove that $f(\bar{x}) = c$ is the same as to prove that $f(\bar{x}) \leq c$. This is immediately given by lower semicontinuity of f : let us choose $x' \in \text{ri Dom } f$ and look what happens along a sequence of points $x_i \in [x', \bar{x})$ converging to \bar{x} . All the points of this sequence are relative interior points of $\text{Dom } f$ (Lemma B.1.1), and consequently

$$f(x_i) = \bar{f}(x_i).$$

Now, $x_i = (1 - \lambda_i)\bar{x} + \lambda_i x'$ with $\lambda_i \rightarrow +0$ as $i \rightarrow \infty$; since \bar{f} clearly is convex (as the upper bound of a family of affine and therefore convex functions), we have

$$\bar{f}(x_i) \leq (1 - \lambda_i)\bar{f}(\bar{x}) + \lambda_i \bar{f}(x').$$

Putting things together, we get

$$f(x_i) \leq (1 - \lambda_i)\bar{f}(\bar{x}) + \lambda_i f(x');$$

as $i \rightarrow \infty$, $x_i \rightarrow \bar{x}$, and the right hand side in our inequality converges to $\bar{f}(\bar{x}) = c$; since f is lower semicontinuous, we get $f(\bar{x}) \leq c$. \blacksquare

We see why “translation of mathematical facts from one mathematical language to another” – in our case, from the language of convex sets to the language of convex functions – may be fruitful: because we invest a lot into the process rather than run it mechanically.

Closure of a convex function. We got a nice result on the “outer description” of a proper convex function: it is the upper bound of a family of affine functions. Note that, vice versa, the upper bound of every family of affine functions is a proper function, provided that this upper bound is finite at least at one point (indeed, as we know from Section C.2.1, upper bound of every family of convex functions is convex, provided that it is finite at least at one point; and Corollary C.6.1 says that upper bound of lower semicontinuous functions (e.g., affine ones – they are even continuous) is lower semicontinuous).

Now, what to do with a convex function which is not lower semicontinuous? The similar question about convex sets – what to do with a convex set which is not closed – can be resolved very simply: we can pass from the set to its closure and thus get a “normal” object which is very “close” to the original one: the “main part” of the original set – its relative interior – remains unchanged, and the “correction” adds to the set something relatively small – the relative boundary. The same approach works for convex functions: if a convex function f is not proper (i.e., its epigraph, being convex and nonempty, is not closed), we can “correct” the function – replace it with a new function with the epigraph being the closure of $\text{Epi}(f)$. To justify this approach, we, of course, should be sure that the closure of the epigraph of a convex function is also an epigraph of such a function. This indeed is the case, and to see it, it suffices to note that a set G in \mathbf{R}^{n+1} is the epigraph of a function taking values in $\mathbf{R} \cup \{+\infty\}$ if and only if the intersection of G with every vertical line $\{x = \text{const}, t \in \mathbf{R}\}$ is either empty, or is a closed ray of the form $\{x = \text{const}, t \geq \bar{t} > -\infty\}$. Now, it is absolutely evident that if G is the closure of the epigraph of a function f , that its intersection with a vertical line is either empty, or is a closed ray, or is the entire line (the last case indeed can take place – look at the closure of the epigraph of the function equal to $-\frac{1}{x}$ for $x > 0$ and $+\infty$ for $x \leq 0$). We see that in order to justify our idea of “proper correction” of a convex function we should prove that if f is convex, then the last of the indicated three cases – the intersection of $\text{cl Epi}(f)$ with a vertical line is the entire line – never occurs. This fact evidently is a corollary of the following simple

Proposition C.6.3 *A convex function is below bounded on every bounded subset of \mathbf{R}^n .*

Proof. Without loss of generality we may assume that the domain of the function f is full-dimensional and that 0 is the interior point of the domain. According to Theorem C.4.1, there exists a neighbourhood U of the origin – which can be thought of to be a centered at the origin ball of some radius $r > 0$ – where f is bounded from above by some C . Now, if $R > 0$ is arbitrary and x is an arbitrary point with $|x| \leq R$, then the point

$$y = -\frac{r}{R}x$$

belongs to U , and we have

$$0 = \frac{r}{r+R}x + \frac{R}{r+R}y;$$

since f is convex, we conclude that

$$f(0) \leq \frac{r}{r+R}f(x) + \frac{R}{r+R}f(y) \leq \frac{r}{r+R}f(x) + \frac{R}{r+R}c,$$

and we get the lower bound

$$f(x) \geq \frac{r+R}{r}f(0) - \frac{r}{R}c$$

for the values of f in the centered at 0 ball of radius R . ■

Thus, we conclude that the closure of the epigraph of a convex function f is the epigraph of certain function, let it be called *the closure $\text{cl } f$ of f* . Of course, this latter function is convex (its epigraph is convex – it is the closure of a convex set), and since its epigraph is closed, $\text{cl } f$ is proper. The following statement gives direct description of $\text{cl } f$ in terms of f :

Proposition C.6.4 *Let f be a convex function and $\text{cl } f$ be its closure. Then*

(i) *For every x one has*

$$\text{cl } f(x) = \lim_{r \rightarrow +0} \inf_{x': \|x' - x\|_2 \leq r} f(x').$$

In particular,

$$f(x) \geq \text{cl } f(x)$$

for all x , and

$$f(x) = \text{cl } f(x)$$

whenever $x \in \text{ri Dom } f$, same as whenever $x \notin \text{cl Dom } f$.

Thus, the “correction” $f \mapsto \text{cl } f$ may vary f only at the points from the relative boundary of $\text{Dom } f$,

$$\text{Dom } f \subset \text{Dom cl } f \subset \text{cl Dom } f,$$

whence also

$$\text{ri Dom } f = \text{ri Dom cl } f.$$

(ii) *The family of affine minorants of $\text{cl } f$ is exactly the family of affine minorants of f , so that*

$$\text{cl } f(x) = \sup\{\phi(x) : \phi \text{ is an affine minorant of } f\},$$

and the sup in the right hand side can be replaced with max whenever $x \in \text{ri Dom cl } f = \text{ri Dom } f$.

[“so that” comes from the fact that $\text{cl } f$ is proper and is therefore the upper bound of its affine minorants]

C.6.2 Subgradients

Let f be a convex function, and let $x \in \text{Dom } f$. It may happen that there exists an affine minorant $d^T x - a$ of f which coincides with f at x :

$$f(y) \geq d^T y - a \quad \forall y, \quad f(x) = d^T x - a.$$

From the equality in the latter relation we get $a = d^T x - f(x)$, and substituting this representation of a into the first inequality, we get

$$f(y) \geq f(x) + d^T(y - x) \quad \forall y. \tag{C.6.3}$$

Thus, if f admits an affine minorant which is exact at x , then there exists d which gives rise to inequality (C.6.3). Vice versa, if d is such that (C.6.3) takes place, then the right hand side of (C.6.3), regarded as a function of y , is an affine minorant of f which is exact at x .

Now note that (C.6.3) expresses certain property of a vector d . A vector satisfying, for a given x , this property – i.e., the slope of an exact at x affine minorant of f – is called a *subgradient* of f at x , and the set of all subgradients of f at x is denoted $\partial f(x)$.

Subgradients of convex functions play important role in the theory and numerical methods of Convex Programming – they are quite reasonable surrogates of gradients. The most elementary properties of the subgradients are summarized in the following statement:

Proposition C.6.5 *Let f be a convex function and x be a point from $\text{Dom } f$. Then*

(i) *$\partial f(x)$ is a closed convex set which for sure is nonempty when $x \in \text{ri Dom } f$*

(ii) *If $x \in \text{int Dom } f$ and f is differentiable at x , then $\partial f(x)$ is the singleton comprised of the usual gradient of f at x .*

Proof. (i): Closedness and convexity of $\partial f(x)$ are evident – (C.6.3) is an infinite system of nonstrict linear inequalities with respect to d , the inequalities being indexed by $y \in \mathbf{R}^n$. Nonemptiness of $\partial f(x)$ for the case when $x \in \text{ri Dom } f$ – this is the most important fact about the subgradients – is readily given by our preceding results. Indeed, we should prove that if $x \in \text{ri Dom } f$, then there exists an affine minorant of f which is exact at x . But this is an immediate consequence of Proposition C.6.4: part (i) of the proposition says that there exists an affine minorant of f which is equal to $\text{cl } f(x)$ at the point x , and part (i) says that $f(x) = \text{cl } f(x)$.

(ii): If $x \in \text{int Dom } f$ and f is differentiable at x , then $\nabla f(x) \in \partial f(x)$ by the Gradient Inequality. To prove that in the case in question $\nabla f(x)$ is the only subgradient of f at x , note that if $d \in \partial f(x)$, then, by definition,

$$f(y) - f(x) \geq d^T(y - x) \quad \forall y$$

Substituting $y - x = th$, h being a fixed direction and t being > 0 , dividing both sides of the resulting inequality by t and passing to limit as $t \rightarrow +0$, we get

$$h^T \nabla f(x) \geq h^T d.$$

This inequality should be valid for all h , which is possible if and only if $d = \nabla f(x)$. ■

Proposition C.6.5 explains why subgradients are good surrogates of gradients: at a point where gradient exists, it is the only subgradient, but, in contrast to the gradient, a subgradient exists basically everywhere (for sure in the relative interior of the domain of the function). E.g., let us look at the simple function

$$f(x) = |x|$$

on the axis. It is, of course, convex (as maximum of two linear forms x and $-x$). Whenever $x \neq 0$, f is differentiable at x with the derivative $+1$ for $x > 0$ and -1 for $x < 0$. At the point $x = 0$ f is not differentiable; nevertheless, it must have subgradients at this point (since 0 is an interior point of the domain of the function). And indeed, it is immediately seen that the subgradients of $|x|$ at $x = 0$ are exactly the reals from the segment $[-1, 1]$. Thus,

$$\partial|x| = \begin{cases} \{-1\}, & x < 0 \\ [-1, 1], & x = 0 \\ \{+1\}, & x > 0 \end{cases}$$

Note also that if x is a relative boundary point of the domain of a convex function, even a “good” one, the set of subgradients of f at x may be empty, as it is the case with the function

$$f(y) = \begin{cases} -\sqrt{y}, & y \geq 0 \\ +\infty, & y < 0 \end{cases};$$

it is clear that there is no non-vertical supporting line to the epigraph of the function at the point $(0, f(0))$, and, consequently, there is no affine minorant of the function which is exact at $x = 0$.

A significant – and important – part of Convex Analysis deals with *subgradient calculus* – with the rules for computing subgradients of “composite” functions, like sums, superpositions, maxima, etc., given subgradients of the operands. These rules extend onto nonsmooth convex case the standard Calculus rules and are very nice and instructive; the related considerations, however, are beyond our scope.

C.6.3 Legendre transformation

Let f be a convex function. We know that f “basically” is the upper bound of all its affine minorants; this is exactly the case when f is proper, otherwise the corresponding equality takes place everywhere except, perhaps, some points from the relative boundary of $\text{Dom } f$. Now, when an affine function $d^T x - a$ is an affine minorant of f ? It is the case if and only if

$$f(x) \geq d^T x - a$$

for all x or, which is the same, if and only if

$$a \geq d^T x - f(x)$$

for all x . We see that if the slope d of an affine function $d^T x - a$ is fixed, then in order for the function to be a minorant of f we should have

$$a \geq \sup_{x \in \mathbf{R}^n} [d^T x - f(x)].$$

The supremum in the right hand side of the latter relation is certain function of d ; this function is called the *Legendre transformation* of f and is denoted f^* :

$$f^*(d) = \sup_{x \in \mathbf{R}^n} [d^T x - f(x)].$$

Geometrically, the Legendre transformation answers the following question: given a slope d of an affine function, i.e., given the hyperplane $t = d^T x$ in \mathbf{R}^{n+1} , what is the minimal “shift down” of the hyperplane which places it below the graph of f ?

From the definition of the Legendre transformation it follows that this is a proper function. Indeed, we loose nothing when replacing $\sup_{x \in \mathbf{R}^n} [d^T x - f(x)]$ by $\sup_{x \in \text{Dom } f} [d^T x - f(x)]$, so that

the Legendre transformation is the upper bound of a family of affine functions. Since this bound is finite at least at one point (namely, at every d coming form affine minorant of f ; we know that such a minorant exists), it is a convex lower semicontinuous function, as claimed.

The most elementary (and the most fundamental) fact about the Legendre transformation is its symmetry:

Proposition C.6.6 *Let f be a convex function. Then twice taken Legendre transformation of f is the closure $\text{cl } f$ of f :*

$$(f^*)^* = \text{cl } f.$$

In particular, if f is proper, then it is the Legendre transformation of its Legendre transformation (which also is proper).

Proof is immediate. The Legendre transformation of f^* at the point x is, by definition,

$$\sup_{d \in \mathbf{R}^n} [x^T d - f^*(d)] = \sup_{d \in \mathbf{R}^n, a \geq f^*(d)} [d^T x - a];$$

the second sup here is exactly the supremum of all affine minorants of f (this is the origin of the Legendre transformation: $a \geq f^*(d)$ if and only if the affine form $d^T x - a$ is a minorant of f). And we already know that the upper bound of all affine minorants of f is the closure of f . ■

The Legendre transformation is a very powerful tool – this is a “global” transformation, so that *local* properties of f^* correspond to *global* properties of f . E.g.,

- $d = 0$ belongs to the domain of f^* if and only if f is below bounded, and if it is the case, then $f^*(0) = -\inf f$;
- if f is proper, then the subgradients of f^* at $d = 0$ are exactly the minimizers of f on \mathbf{R}^n ;
- $\text{Dom } f^*$ is the entire \mathbf{R}^n if and only if $f(x)$ grows, as $\|x\|_2 \rightarrow \infty$, faster than $\|x\|_2$: there exists a function $r(t) \rightarrow \infty$, as $t \rightarrow \infty$ such that

$$f(x) \geq r(\|x\|_2) \quad \forall x,$$

etc. Thus, whenever we can compute explicitly the Legendre transformation of f , we get a lot of “global” information on f . Unfortunately, the more detailed investigation of the properties of Legendre transformation is beyond our scope; I simply list several simple facts and examples:

- From the definition of Legendre transformation,

$$f(x) + f^*(d) \geq x^T d \quad \forall x, d.$$

Specifying here f and f^* , we get certain inequality, e.g., the following one:

[Young’s Inequality] if p and q are positive reals such that $\frac{1}{p} + \frac{1}{q} = 1$, then

$$\frac{|x|^p}{p} + \frac{|d|^q}{q} \geq xd \quad \forall x, d \in \mathbf{R}$$

(indeed, as it is immediately seen, the Legendre transformation of the function $|x|^p/p$ is $|d|^q/q$)

Consequences. Very simple-looking Young’s inequality gives rise to a very nice and useful Hölder inequality:

Let $1 \leq p \leq \infty$ and let q be such $\frac{1}{p} + \frac{1}{q} = 1$ ($p = 1 \Rightarrow q = \infty$, $p = \infty \Rightarrow q = 1$). For every two vectors $x, y \in \mathbf{R}^n$ one has

$$\sum_{i=1}^n |x_i y_i| \leq \|x\|_p \|y\|_q \tag{C.6.4}$$

Indeed, there is nothing to prove if p or q is ∞ – if it is the case, the inequality becomes the evident relation

$$\sum_i |x_i y_i| \leq (\max_i |x_i|) (\sum_i |y_i|).$$

Now let $1 < p < \infty$, so that also $1 < q < \infty$. In this case we should prove that

$$\sum_i |x_i y_i| \leq (\sum_i |x_i|^p)^{1/p} (\sum_i |y_i|^q)^{1/q}.$$

There is nothing to prove if one of the factors in the right hand side vanishes; thus, we can assume that $x \neq 0$ and $y \neq 0$. Now, both sides of the inequality are of homogeneity degree 1 with respect to x (when we multiply x by t , both sides are multiplied by $|t|$), and similarly with respect to y . Multiplying x and y by appropriate reals, we can make both factors in the right hand side equal to 1: $\|x\|_p = \|y\|_q = 1$. Now we should prove that under this normalization the left hand side in the inequality is ≤ 1 , which is immediately given by the Young inequality:

$$\sum_i |x_i y_i| \leq \sum_i [|x_i|^p/p + |y_i|^q/q] = 1/p + 1/q = 1.$$

Note that the Hölder inequality says that

$$|x^T y| \leq \|x\|_p \|y\|_q; \quad (\text{C.6.5})$$

when $p = q = 2$, we get the Cauchy inequality. Now, inequality (C.6.5) is *exact* in the sense that for every x there exists y with $\|y\|_q = 1$ such that

$$x^T y = \|x\|_p \quad [= \|x\|_p \|y\|_q];$$

it suffices to take

$$y_i = \|x\|_p^{1-p} |x_i|^{p-1} \operatorname{sign}(x_i)$$

(here $x \neq 0$; the case of $x = 0$ is trivial – here y can be an arbitrary vector with $\|y\|_q = 1$).

Combining our observations, we come to an extremely important, although simple, fact:

$$\|x\|_p = \max\{y^T x : \|y\|_q \leq 1\} \quad \left[\frac{1}{p} + \frac{1}{q} = 1\right]. \quad (\text{C.6.6})$$

It follows, in particular, that $\|x\|_p$ is convex (as an upper bound of a family of linear forms), whence

$$\|x' + x''\|_p = 2\left\|\frac{1}{2}x' + \frac{1}{2}x''\right\|_p \leq 2(\|x'\|_p/2 + \|x''\|_p/2) = \|x'\|_p + \|x''\|_p;$$

this is nothing but the triangle inequality. Thus, $\|x\|_p$ satisfies the triangle inequality; it clearly possesses two other characteristic properties of a norm – positivity and homogeneity. Consequently, $\|\cdot\|_p$ is a norm – the fact that we announced twice and have finally proven now.

- The Legendre transformation of the function

$$f(x) \equiv -a$$

is the function which is equal to a at the origin and is $+\infty$ outside the origin; similarly, the Legendre transformation of an affine function $\bar{d}^T x - a$ is equal to a at $d = \bar{d}$ and is $+\infty$ when $d \neq \bar{d}$;

- The Legendre transformation of the strictly convex quadratic form

$$f(x) = \frac{1}{2}x^T Ax$$

(A is positive definite symmetric matrix) is the quadratic form

$$f^*(d) = \frac{1}{2}d^T A^{-1}d$$

- The Legendre transformation of the Euclidean norm

$$f(x) = \|x\|_2$$

is the function which is equal to 0 in the closed unit ball centered at the origin and is $+\infty$ outside the ball.

The latter example is a particular case of the following statement:

Let $\|x\|$ be a norm on \mathbf{R}^n , and let

$$\|d\|_* = \sup\{d^T x : \|x\| \leq 1\}$$

be the conjugate to $\|\cdot\|$ norm.

Exercise C.1 Prove that $\|\cdot\|_*$ is a norm, and that the norm conjugate to $\|\cdot\|_*$ is the original norm $\|\cdot\|$.

Hint: Observe that the unit ball of $\|\cdot\|_*$ is exactly the polar of the unit ball of $\|\cdot\|$.

The Legendre transformation of $\|x\|$ is the characteristic function of the unit ball of the conjugate norm, i.e., is the function of d equal to 0 when $\|d\|_* \leq 1$ and is $+\infty$ otherwise.

E.g., (C.6.6) says that the norm conjugate to $\|\cdot\|_p$, $1 \leq p \leq \infty$, is $\|\cdot\|_q$, $1/p + 1/q = 1$; consequently, the Legendre transformation of p -norm is the characteristic function of the unit $\|\cdot\|_q$ -ball.

Appendix D

Convex Programming, Lagrange Duality, Saddle Points

D.1 Mathematical Programming Program

A (constrained) Mathematical Programming program is a problem as follows:

$$(P) \quad \min \{f(x) : x \in X, \quad g(x) \equiv (g_1(x), \dots, g_m(x)) \leq 0, \quad h(x) \equiv (h_1(x), \dots, h_k(x)) = 0\}. \quad (\text{D.1.1})$$

The standard terminology related to (D.1.1) is:

- [domain] X is called the *domain* of the problem
- [objective] f is called the *objective*
- [constraints] $g_i, i = 1, \dots, m$, are called the (functional) *inequality constraints*; $h_j, j = 1, \dots, k$, are called the *equality constraints*¹⁾

In the sequel, if the opposite is not explicitly stated, it always is assumed that the objective and the constraints are well-defined on X .

- [feasible solution] a point $x \in \mathbf{R}^n$ is called a *feasible solution* to (D.1.1), if $x \in X, g_i(x) \leq 0, i = 1, \dots, m$, and $h_j(x) = 0, j = 1, \dots, k$, i.e., if x satisfies all restrictions imposed by the formulation of the problem
 - [feasible set] the set of all feasible solutions is called the *feasible set* of the problem
 - [feasible problem] a problem with a nonempty feasible set (i.e., the one which admits feasible solutions) is called *feasible* (or consistent)
 - [active constraints] an inequality constraint $g_i(\cdot) \leq 0$ is called *active at a given feasible solution* x , if this constraint is satisfied at the point as an equality rather than strict inequality, i.e., if

$$g_i(x) = 0.$$

A equality constraint $h_i(x) = 0$ by definition is active at every feasible solution x .

¹⁾rigorously speaking, the constraints are not the functions g_i, h_j , but the relations $g_i(x) \leq 0, h_j(x) = 0$; in fact the word “constraints” is used in both these senses, and it is always clear what is meant. E.g., saying that x satisfies the constraints, we mean the relations, and saying that the constraints are differentiable, we mean the functions

- [optimal value] the quantity

$$f^* = \begin{cases} \inf_{x \in X: g(x) \leq 0, h(x)=0} f(x), & \text{the problem is feasible} \\ +\infty, & \text{the problem is infeasible} \end{cases}$$

is called *the optimal value* of the problem

- [below boundedness] the problem is called *below bounded*, if its optimal value is $> -\infty$, i.e., if the objective is below bounded on the feasible set
- [optimal solution] a point $x \in \mathbf{R}^n$ is called an *optimal solution* to (D.1.1), if x is feasible and $f(x) \leq f(x')$ for any other feasible solution, i.e., if

$$x \in \operatorname{Argmin}_{x' \in X: g(x') \leq 0, h(x')=0} f(x')$$

- [solvable problem] a problem is called *solvable*, if it admits optimal solutions
- [optimal set] the set of all optimal solutions to a problem is called its *optimal set*

To solve the problem *exactly* means to find its optimal solution or to detect that no optimal solution exists.

D.2 Convex Programming program and Lagrange Duality Theorem

A Mathematical Programming program (P) is called *convex* (or *Convex Programming* program), if

- X is a convex subset of \mathbf{R}^n
- f, g_1, \dots, g_m are *real-valued convex functions* on X ,
- and
- there are no equality constraints at all.

Note that instead of saying that there are no equality constraints, we could say that there are constraints of this type, but only *linear* ones; this latter case can be immediately reduced to the one without equality constraints by replacing \mathbf{R}^n with the affine subspace given by the (linear) equality constraints.

D.2.1 Convex Theorem on Alternative

The simplest case of a convex program is, of course, a Linear Programming program – the one where $X = \mathbf{R}^n$ and the objective and all the constraints are linear. We already know what are optimality conditions for this particular case – they are given by the Linear Programming Duality Theorem. How did we get these conditions?

We started with the observation that the fact that a point x^* is an optimal solution can be expressed in terms of solvability/unsolvability of certain systems of inequalities: in our now terms, these systems are

$$x \in G, f(x) \leq c, g_j(x) \leq 0, j = 1, \dots, m \quad (\text{D.2.1})$$

and

$$x \in G, f(x) < c, g_j(x) \leq 0, j = 1, \dots, m; \quad (\text{D.2.2})$$

here c is a parameter. Optimality of x^* for the problem means exactly that for appropriately chosen c (this choice, of course, is $c = f(x^*)$) the first of these systems is solvable and x^* is its solution, while the second system is unsolvable. Given this trivial observation, we converted the “negative” part of it – the claim that (D.2.2) is unsolvable – into a positive statement, using the General Theorem on Alternative, and this gave us the LP Duality Theorem.

Now we are going to use the same approach. What we need is a “convex analogy” to the Theorem on Alternative – something like the latter statement, but for the case when the inequalities in question are given by convex functions rather than the linear ones (and, besides it, we have a “convex inclusion” $x \in X$).

It is easy to guess the result we need. How did we come to the formulation of the Theorem on Alternative? The question we were interested in was, basically, how to express in an affirmative manner the fact that a system of linear inequalities has no solutions; to this end we observed that if we can combine, in a linear fashion, the inequalities of the system and get an obviously false inequality like $0 \leq -1$, then the system is unsolvable; this condition is certain affirmative statement with respect to the weights with which we are combining the original inequalities.

Now, the scheme of the above reasoning has nothing in common with linearity (and even convexity) of the inequalities in question. Indeed, consider an arbitrary inequality system of the type (D.2.2):

$$\begin{aligned} & (I) \\ & f(x) < c \\ & g_j(x) \leq 0, j = 1, \dots, m \\ & x \in X; \end{aligned}$$

all we assume is that X is a nonempty subset in \mathbf{R}^n and f, g_1, \dots, g_m are real-valued functions on X . It is absolutely evident that

if there exist nonnegative $\lambda_1, \dots, \lambda_m$ such that the inequality

$$f(x) + \sum_{j=1}^m \lambda_j g_j(x) < c \quad (\text{D.2.3})$$

has no solutions in X , then (I) also has no solutions.

Indeed, a solution to (I) clearly is a solution to (D.2.3) – the latter inequality is nothing but a combination of the inequalities from (I) with the weights 1 (for the first inequality) and λ_j (for the remaining ones).

Now, what does it mean that (D.2.3) has no solutions? A necessary and sufficient condition for this is that the infimum of the left hand side of (D.2.3) in $x \in X$ is $\geq c$. Thus, we come to the following evident

Proposition D.2.1 [Sufficient condition for insolvability of (I)] Consider a system (I) with arbitrary data and assume that the system

$$(II) \quad \begin{aligned} \inf_{x \in X} \left[f(x) + \sum_{j=1}^m \lambda_j g_j(x) \right] &\geq c \\ \lambda_j &\geq 0, j = 1, \dots, m \end{aligned}$$

with unknowns $\lambda_1, \dots, \lambda_m$ has a solution. Then (I) is infeasible.

Let me stress that this result is completely general; it does not require any assumptions on the entities involved.

The result we have obtained, unfortunately, does not help us: the actual power of the Theorem on Alternative (and the fact used to prove the Linear Programming Duality Theorem) is not the *sufficiency* of the condition of Proposition for infeasibility of (I), but the *necessity* of this condition. Justification of necessity of the condition in question has nothing in common with the evident reasoning which gives the sufficiency. The necessity in the linear case ($X = \mathbf{R}^n$, f , g_1, \dots, g_m are linear) can be established via the Homogeneous Farkas Lemma. Now we shall prove the necessity of the condition for the convex case, and already here we need some additional, although minor, assumptions; and in the general nonconvex case the condition in question simply is *not* necessary for infeasibility of (I) [and this is very bad – this is the reason why there exist difficult optimization problems which we do not know how to solve efficiently].

The just presented “preface” explains what we should do; now let us carry out our plan. We start with the aforementioned “minor regularity assumptions”.

Definition D.2.1 [Slater Condition] Let $X \subset \mathbf{R}^n$ and g_1, \dots, g_m be real-valued functions on X . We say that these functions satisfy the Slater condition on X , if there exists $x \in X$ such that $g_j(x) < 0$, $j = 1, \dots, m$.

An inequality constrained program

$$(IC) \quad \min \{f(x) : g_j(x) \leq 0, j = 1, \dots, m, x \in X\}$$

(f, g_1, \dots, g_m are real-valued functions on X) is called to satisfy the Slater condition, if g_1, \dots, g_m satisfy this condition on X .

We are about to establish the following fundamental fact:

Theorem D.2.1 [Convex Theorem on Alternative]

Let $X \subset \mathbf{R}^n$ be convex, let f, g_1, \dots, g_m be real-valued convex functions on X , and let g_1, \dots, g_m satisfy the Slater condition on X . Then system (I) is solvable if and only if system (II) is unsolvable.

Proof. The first part of the statement – “if (II) has a solution, then (I) has no solutions” – is given by Proposition D.2.1. What we need is to prove the inverse statement. Thus, let us assume that (I) has no solutions, and let us prove that then (II) has a solution.

Without loss of generality we may assume that X is full-dimensional: $\text{ri } X = \text{int } X$ (indeed, otherwise we could replace our “universe” \mathbf{R}^n with the affine span of X).

¹⁰. Let us set

$$F(x) = \begin{pmatrix} f(x) \\ g_1(x) \\ \dots \\ g_m(x) \end{pmatrix}$$

and consider two sets in \mathbf{R}^{m+1} :

$$S = \{u = (u_0, \dots, u_m) \mid \exists x \in X : F(x) \leq u\}$$

and

$$T = \{(u_0, \dots, u_m) \mid u_0 < c, u_1 \leq 0, u_2 \leq 0, \dots, u_m \leq 0\}.$$

I claim that

- (i) S and T are nonempty convex sets;
- (ii) S and T does not intersect.

Indeed, convexity and nonemptiness of T is evident, same as nonemptiness of S . Convexity of S is an immediate consequence of the fact that X and f, g_1, \dots, g_m are convex. Indeed, assuming that $u', u'' \in S$, we conclude that there exist $x', x'' \in X$ such that $F(x') \leq u'$ and $F(x'') \leq u''$, whence, for every $\lambda \in [0, 1]$.

$$\lambda F(x') + (1 - \lambda)F(x'') \leq \lambda u' + (1 - \lambda)u''.$$

The left hand side in this inequality, due to convexity of X and f, g_1, \dots, g_m , is $\geq F(y)$, $y = \lambda x' + (1 - \lambda)x''$. Thus, for the point $v = \lambda u' + (1 - \lambda)u''$ there exists $y \in X$ with $F(y) \leq v$, whence $v \in S$. Thus, S is convex.

The fact that $S \cap T = \emptyset$ is an evident equivalent reformulation of the fact that (I) has no solutions.

2⁰. Since S and T are nonempty convex sets with empty intersection, by Separation Theorem (Theorem B.2.9) they can be separated by a linear form: there exist $a = (a_0, \dots, a_m) \neq 0$ such that

$$\inf_{u \in S} \sum_{j=0}^m a_j u_j \geq \sup_{u \in T} \sum_{j=0}^m a_j u_j. \quad (\text{D.2.4})$$

3⁰. Let us look what can be said about the vector a . I claim that, first,

$$a \geq 0 \quad (\text{D.2.5})$$

and, second,

$$a_0 > 0. \quad (\text{D.2.6})$$

Indeed, to prove (D.2.5) note that if some a_i were negative, then the right hand side in (D.2.4) would be $+\infty$ ²⁾, which is forbidden by (D.2.4).

Thus, $a \geq 0$; with this in mind, we can immediately compute the right hand side of (D.2.4):

$$\sup_{u \in T} \sum_{j=0}^m a_j u_j = \sup_{u_0 < c, u_1, \dots, u_m \leq 0} \sum_{j=0}^m a_j u_j = a_0 c.$$

Since for every $x \in X$ the point $F(x)$ belongs to S , the left hand side in (D.2.4) is not greater than

$$\inf_{x \in X} \left[a_0 f(x) + \sum_{j=1}^m a_j g_j(x) \right];$$

²⁾look what happens when all coordinates in u , except the i th one, are fixed at values allowed by the description of T and u_i is a large in absolute value negative real

combining our observations, we conclude that (D.2.4) implies the inequality

$$\inf_{x \in X} \left[a_0 f(x) + \sum_{j=1}^m a_j g_j(x) \right] \geq a_0 c. \quad (\text{D.2.7})$$

Now let us prove that $a_0 > 0$. This crucial fact is an immediate consequence of the Slater condition. Indeed, let $\bar{x} \in X$ be the point given by this condition, so that $g_j(\bar{x}) < 0$. From (D.2.7) we conclude that

$$a_0 f(\bar{x}) + \sum_{j=0}^m a_j g_j(\bar{x}) \geq a_0 c.$$

If a_0 were 0, then the right hand side of this inequality would be 0, while the left one would be the combination $\sum_{j=0}^m a_j g_j(\bar{x})$ of negative reals $g_j(\bar{x})$ with nonnegative coefficients a_j not all equal to 0³⁾, so that the left hand side is strictly negative, which is the desired contradiction.

⁴⁾ Now we are done: since $a_0 > 0$, we are in our right to divide both sides of (D.2.7) by a_0 and thus get

$$\inf_{x \in X} \left[f_0(x) + \sum_{j=1}^m \lambda_j g_j(x) \right] \geq c, \quad (\text{D.2.8})$$

where $\lambda_j = a_j/a_0 \geq 0$. Thus, (II) has a solution. ■

D.2.2 Lagrange Function and Lagrange Duality

D.2.2.1 D.2.2.A. Lagrange function

The result of Convex Theorem on Alternative brings to our attention the function

$$\underline{L}(\lambda) = \inf_{x \in X} \left[f(x) + \sum_{j=1}^m \lambda_j g_j(x) \right], \quad (\text{D.2.9})$$

same as the aggregate

$$L(x, \lambda) = f(x) + \sum_{j=1}^m \lambda_j g_j(x) \quad (\text{D.2.10})$$

from which this function comes. Aggregate (D.2.10) is called the *Lagrange function* of the inequality constrained optimization program

$$(IC) \quad \min \{f(x) : g_j(x) \leq 0, j = 1, \dots, m, x \in X\}.$$

The Lagrange function of an optimization program is a very important entity: most of optimality conditions are expressed in terms of this function. Let us start with translating of what we already know to the language of the Lagrange function.

³⁾indeed, from the very beginning we know that $a \neq 0$, so that if $a_0 = 0$, then not all a_j , $j \geq 1$, are zeros

D.2.2.2 D.2.2.B. Convex Programming Duality Theorem

Theorem D.2.2 Consider an arbitrary inequality constrained optimization program (IC). Then

- (i) The infimum

$$\underline{L}(\lambda) = \inf_{x \in X} L(x, \lambda)$$

of the Lagrange function in $x \in X$ is, for every $\lambda \geq 0$, a lower bound on the optimal value in (IC), so that the optimal value in the optimization program

$$(IC^*) \quad \sup_{\lambda \geq 0} \underline{L}(\lambda)$$

also is a lower bound for the optimal value in (IC);

- (ii) [Convex Duality Theorem] If (IC)

- is convex,
- is below bounded

and

- satisfies the Slater condition,

then the optimal value in (IC^*) is attained and is equal to the optimal value in (IC).

Proof. (i) is nothing but Proposition D.2.1 (why?). It makes sense, however, to repeat here the corresponding one-line reasoning:

Let $\lambda \geq 0$; in order to prove that

$$\underline{L}(\lambda) \equiv \inf_{x \in X} L(x, \lambda) \leq c^* \quad [L(x, \lambda) = f(x) + \sum_{j=1}^m \lambda_j g_j(x)],$$

where c^* is the optimal value in (IC), note that if x is feasible for (IC), then evidently $L(x, \lambda) \leq f(x)$, so that the infimum of L over $x \in X$ is \leq the infimum c^* of f over the feasible set of (IC). \square

(ii) is an immediate consequence of the Convex Theorem on Alternative. Indeed, let c^* be the optimal value in (IC). Then the system

$$f(x) < c^*, g_j(x) \leq 0, j = 1, \dots, m$$

has no solutions in X , and by the above Theorem the system (II) associated with $c = c^*$ has a solution, i.e., there exists $\lambda^* \geq 0$ such that $\underline{L}(\lambda^*) \geq c^*$. But we know from (i) that the strict inequality here is impossible and, besides this, that $\underline{L}(\lambda) \leq c^*$ for every $\lambda \geq 0$. Thus, $\underline{L}(\lambda^*) = c^*$ and λ^* is a maximizer of \underline{L} over $\lambda \geq 0$. \blacksquare

D.2.2.3 C.1.2.C. Dual program

Theorem D.2.2 establishes certain connection between two optimization programs – the “primal” program

$$(IC) \quad \min \{f(x) : g_j(x) \leq 0, j = 1, \dots, m, x \in X\}$$

and its *Lagrange dual program*

$$(IC^*) \quad \max \left\{ \underline{L}(\lambda) \equiv \inf_{x \in X} L(x, \lambda) : \lambda \geq 0 \right\}$$

(the variables λ of the dual problem are called the *Lagrange multipliers* of the primal problem). The Theorem says that the optimal value in the dual problem is \leq the one in the primal, and under some favourable circumstances (the primal problem is convex below bounded and satisfies the Slater condition) the optimal values in the programs are equal to each other.

In our formulation there is some asymmetry between the primal and the dual programs. In fact both of the programs are related to the Lagrange function in a quite symmetric way. Indeed, consider the program

$$\min_{x \in X} \bar{L}(x), \quad \bar{L}(x) = \sup_{\lambda \geq 0} L(x, \lambda).$$

The objective in this program clearly is $+\infty$ at every point $x \in X$ which is not feasible for (IC) and is $f(x)$ on the feasible set of (IC), so that the program is equivalent to (IC). We see that both the primal and the dual programs come from the Lagrange function: in the primal problem, we minimize over X the result of maximization of $L(x, \lambda)$ in $\lambda \geq 0$, and in the dual program we maximize over $\lambda \geq 0$ the result of minimization of $L(x, \lambda)$ in $x \in X$. This is a particular (and the most important) example of a *zero sum two person game* – the issue we will speak about later.

We have seen that under certain convexity and regularity assumptions the optimal values in (IC) and (IC^{*}) are equal to each. There is also another way to say when these optimal values are equal – this is always the case when the Lagrange function possesses a saddle point, i.e., there exists a pair $x^* \in X, \lambda^* \geq 0$ such that at the pair $L(x, \lambda)$ attains its minimum as a function of $x \in X$ and attains its maximum as a function of $\lambda \geq 0$:

$$L(x, \lambda^*) \geq L(x^*, \lambda^*) \geq L(x^*, \lambda) \quad \forall x \in X, \lambda \geq 0.$$

It can be easily demonstrated (do it by yourself or look at Theorem D.4.1) that

Proposition D.2.2 (x^*, λ^*) is a saddle point of the Lagrange function L of (IC) if and only if x^* is an optimal solution to (IC), λ^* is an optimal solution to (IC^{*}) and the optimal values in the indicated problems are equal to each other.

Our current goal is to extract from what we already know optimality conditions for convex programs.

D.2.3 Optimality Conditions in Convex Programming

D.2.3.1 D.2.3.A. Saddle point form of optimality conditions

Theorem D.2.3 [Saddle Point formulation of Optimality Conditions in Convex Programming] Let (IC) be an optimization program, $L(x, \lambda)$ be its Lagrange function, and let $x^* \in X$. Then

(i) A sufficient condition for x^* to be an optimal solution to (IC) is the existence of the vector of Lagrange multipliers $\lambda^* \geq 0$ such that (x^*, λ^*) is a saddle point of the Lagrange function $L(x, \lambda)$, i.e., a point where $L(x, \lambda)$ attains its minimum as a function of $x \in X$ and attains its maximum as a function of $\lambda \geq 0$:

$$L(x, \lambda^*) \geq L(x^*, \lambda^*) \geq L(x^*, \lambda) \quad \forall x \in X, \lambda \geq 0. \quad (\text{D.2.11})$$

(ii) if the problem (IC) is convex and satisfies the Slater condition, then the above condition is necessary for optimality of x^* : if x^* is optimal for (IC), then there exists $\lambda^* \geq 0$ such that (x^*, λ^*) is a saddle point of the Lagrange function.

Proof. (i): assume that for a given $x^* \in X$ there exists $\lambda^* \geq 0$ such that (D.2.11) is satisfied, and let us prove that then x^* is optimal for (IC). First of all, x^* is feasible: indeed, if $g_j(x^*) > 0$ for some j , then, of course, $\sup_{\lambda \geq 0} L(x^*, \lambda) = +\infty$ (look what happens when all λ 's, except λ_j , are fixed, and $\lambda_j \rightarrow +\infty$); but $\sup_{\lambda \geq 0} L(x^*, \lambda) = +\infty$ is forbidden by the second inequality in (D.2.11).

Since x^* is feasible, $\sup_{\lambda \geq 0} L(x^*, \lambda) = f(x^*)$, and we conclude from the second inequality in (D.2.11) that $L(x^*, \lambda^*) = f(x^*)$. Now the first inequality in (D.2.11) reads

$$f(x) + \sum_{j=1}^m \lambda_j^* g_j(x) \geq f(x^*) \quad \forall x \in X.$$

This inequality immediately implies that x^* is optimal: indeed, if x is feasible for (IC), then the left hand side in the latter inequality is $\leq f(x)$ (recall that $\lambda^* \geq 0$), and the inequality implies that $f(x) \geq f(x^*)$. \square

(ii): Assume that (IC) is a convex program, x^* is its optimal solution and the problem satisfies the Slater condition; we should prove that then there exists $\lambda^* \geq 0$ such that (x^*, λ^*) is a saddle point of the Lagrange function, i.e., that (D.2.11) is satisfied. As we know from the Convex Programming Duality Theorem (Theorem D.2.2.(ii)), the dual problem (IC^*) has a solution $\lambda^* \geq 0$ and the optimal value of the dual problem is equal to the optimal value in the primal one, i.e., to $f(x^*)$:

$$f(x^*) = \underline{L}(\lambda^*) \equiv \inf_{x \in X} L(x, \lambda^*). \quad (\text{D.2.12})$$

We immediately conclude that

$$\lambda_j^* > 0 \Rightarrow g_j(x^*) = 0$$

(this is called *complementary slackness*: positive Lagrange multipliers can be associated only with active (satisfied at x^* as equalities) constraints. Indeed, from (D.2.12) it follows that

$$f(x^*) \leq L(x^*, \lambda^*) = f(x^*) + \sum_{j=1}^m \lambda_j^* g_j(x^*);$$

the terms in the \sum_j in the right hand side are nonpositive (since x^* is feasible for (IC)), and the sum itself is nonnegative due to our inequality; it is possible if and only if all the terms in the sum are zero, and this is exactly the complementary slackness.

From the complementary slackness we immediately conclude that $f(x^*) = L(x^*, \lambda^*)$, so that (D.2.12) results in

$$L(x^*, \lambda^*) = f(x^*) = \inf_{x \in X} L(x, \lambda^*).$$

On the other hand, since x^* is feasible for (IC), we have $L(x^*, \lambda) \leq f(x^*)$ whenever $\lambda \geq 0$. Combining our observations, we conclude that

$$L(x^*, \lambda) \leq L(x^*, \lambda^*) \leq L(x, \lambda^*)$$

for all $x \in X$ and all $\lambda \geq 0$. ■

Note that (i) is valid for an arbitrary inequality constrained optimization program, not necessarily convex. However, in the nonconvex case the *sufficient* condition for optimality given by (i) is extremely far from being necessary and is “almost never” satisfied. In contrast to this, in the convex case the condition in question is not only sufficient, but also “nearly necessary” – it for sure is necessary when (IC) is a convex program satisfying the Slater condition.

We are about to prove a modification of Theorem D.2.3, where we slightly relax the Slater condition.

Theorem D.2.4 *Consider a convex problem (IC), and let x^* be a feasible solution of the problem. Assume that the functions g_1, \dots, g_k are affine, while the functions f, g_{k+1}, \dots, g_m are differentiable at x . Finally, assume the restricted Slater condition: there exists $\bar{x} \in \text{ri } X$ such that $g_i(\bar{x}) \leq 0$ for $i \leq k$ and $g_i(\bar{x}) < 0$ for $i > k$. Then x^* is an optimal solution to (IC) if and only if there exists $\lambda^* \geq 0$ such that (x^*, λ^*) is a saddle point of $L(x, \lambda)$ on $X \times \{\lambda \geq 0\}$.*

Proof. The “if” part of the statement is given by Theorem D.2.3.(i). Let us focus on the “only if” part. Thus, assume that x^* is an optimal solution of (IC), and let us prove the existence of required λ_* . As always, we may assume without loss of generality that $\text{int } X \neq \emptyset$. Let $I(x^*)$ be the set of indices of the constraints which are active at x^* . Consider the radial cone of X at x^* :

$$M_1 = \{h : \exists t > 0 : x^* + th \in X\}$$

along with the polyhedral cone

$$M_2 = \{h : (\nabla g_j^T(x^*))h \leq 0 \ \forall j \in I(x^*)\}.$$

We claim that

(I): M_2 is a closed cone which has a nonempty intersection with the interior of the convex cone M_1 ;

(II): the vector $\nabla f(x^*)$ belongs to the cone dual to the cone $M = M_1 \cap M_2$.

Postponing for the time being the proofs, let us derive from (I), (II) the existence of the required vector of Lagrange multipliers. Applying the Dubovitskii-Milutin Lemma (Theorem B.2.7), which is legitimate due to (I), (II), we conclude that there exists a representation

$$\nabla f(x^*) = u + v, \quad u \in M'_1, v \in M'_2,$$

where M'_i is the cone dual to the cone M_i . By the Homogeneous Farkas Lemma, we have

$$v = - \sum_{j \in I(x^*)} \lambda_j^* \nabla g_j(x^*),$$

where $\lambda_j^* \geq 0$. Setting $\lambda_j^* = 0$ for $j \notin I(x^*)$, we get a vector $\lambda^* \geq 0$ such that

$$\begin{aligned} \nabla_x \Big|_{x=x^*} L(x, \lambda^*) &= \nabla f(x^*) + \sum_j \lambda_j^* \nabla g_j(x^*) = \nabla f(x^*) - v = u, \\ \lambda_j^* g_j(x^*) &= 0, \quad j = 1, \dots, m. \end{aligned} \tag{D.2.13}$$

Since the function $L(x, \lambda^*)$ is convex in $x \in X$ and differentiable at x^* , the first relation in (D.2.13) combines with the inclusion $u \in M'_1$ and Proposition C.5.1 to imply that x^* is a minimizer of $L(x, \lambda^*)$ over $x \in X$. The second relation in (D.2.13) is the complementary slackness which, as we remember from the proof of Theorem D.2.3.(ii), combines with the feasibility of x^* to imply that λ^* is a maximizer of $L(x^*, \lambda)$ over $\lambda \geq 0$. Thus, (x^*, λ^*) is a saddle point of the Lagrange function, as claimed.

It remains to verify (I) and (II).

(I): the fact that M_2 is a closed cone is evident (M_2 is a polyhedral cone). The fact that M_1 is a convex cone with a nonempty interior is an immediate consequence of the convexity of X and the relation $\text{int } X \neq \emptyset$. By assumption, there exists a point $\bar{x} \in \text{int } X$ such that $g_j(\bar{x}) \leq 0$ for all j . Since $\bar{x} \in \text{int } X$, the vector $h = \bar{x} - x^*$ clearly belongs to $\text{int } M_1$; since $g_j(x^*) = 0$, $j \in I(x^*)$, and $g_j(\bar{x}) \leq 0$, from Gradient Inequality it follows that $h^T \nabla g_j(x^*) \leq g_j(\bar{x}) - g_j(x^*) \leq 0$ for $j \in I(x^*)$, so that $h \in M_1$. Thus, h is the intersection of $\text{int } M_1$ and M_2 , so that this intersection is nonempty. \square

(II): Assume, on the contrary to what should be proven, that there exists a vector $d \in M_1 \cap M_2$ such that $d^T \nabla f(x^*) < 0$. Let h be the same vector as in the proof of (I). Since $d^T \nabla f(x^*) < 0$, we can choose $\epsilon > 0$ such that with $d_\epsilon = d + \epsilon h$ one has $d_\epsilon^T \nabla f(x^*) < 0$. Since both d and h belong to M_1 , there exists $\delta > 0$ such that $x_t = x^* + t d_\epsilon \in X$ for $0 \leq t \leq \delta$; since $d_\epsilon^T \nabla f(x^*) < 0$, we may further assume that $f(x_t) < f(x^*)$ when $0 < t \leq \delta$. Let us verify that for every $j \leq m$ one has

$$(*_j) : \text{There exists } \delta_j > 0 \text{ such that } g_j(x_t) \leq 0 \text{ for } 0 \leq t \leq \delta_j.$$

This will yield the desired contradiction, since, setting $t = \min[\delta, \min_j \delta_j]$, we would have $x_t \in X$, $g_j(x_t) \leq 0$, $j = 1, \dots, m$, $f(x_t) < f(x^*)$, which is impossible, since x^* is an optimal solution of (IC).

To prove $(*_j)$, consider the following three possibilities:

$j \notin I(x^*)$: here $(*_j)$ is evident, since $g_j(x)$ is negative at x^* and is continuous in $x \in X$ at the point x^* (recall that all g_j are assumed even to be differentiable at x^*).

$j \in I(x^*)$ and $j \leq k$: For j in question, the function $g_j(x)$ is affine and vanishes at x^* , while $\nabla g_j(x^*)$ has nonpositive inner products with both d (due to $d \in M_2$) and h (due to $g_j(x^*) = 0$, $g_j(x^* + h) = g_j(\bar{x}) \leq 0$); it follows that $\nabla g_j(x^*)$ has nonpositive inner product with d_ϵ , and since the function is affine, we arrive at $g_j(x^* + t d_\epsilon) \leq g_j(x^*) = 0$ for $t \geq 0$.

$j \in I(x^*)$ and $j > k$: In this case, the function $\gamma_j(t) = g_j(x^* + t \epsilon)$ vanishes at $t = 0$ and is differentiable at $t = 0$ with the derivative $\gamma'_j(0) = (\epsilon h + d)^T \nabla g_j(x^*)$. This derivative is negative, since $d^T \nabla g_j(x^*) \leq 0$ due to $d \in M_2$ and $j \in I(x^*)$, while by the Gradient Inequality $h^T \nabla g_j(x^*) \leq g_j(x^* + h) - g_j(x^*) = g_j(\bar{x}) - g_j(x^*) \leq g_j(\bar{x}) < 0$. Since $\gamma_j(0) = 0$, $\gamma'_j(0) < 0$, $\gamma_j(t)$ is negative for all small enough positive t , as required in $(*_j)$. \blacksquare

D.2.3.2 D.2.3.B. Karush-Kuhn-Tucker form of optimality conditions

Theorems D.2.3, D.2.4 express, basically, the strongest optimality conditions for a Convex Programming program. These conditions, however, are “implicit” – they are expressed in terms of saddle point of the Lagrange function, and it is unclear how to verify that something is or is not the saddle point of the Lagrange function. Fortunately, the proof of Theorem D.2.4 yields more or less explicit optimality conditions as follows:

Theorem D.2.5 [Karush-Kuhn-Tucker Optimality Conditions in Convex Programming] *Let (IC) be a convex program, let x^* be its feasible solution, and let the functions f, g_1, \dots, g_m be differentiable at x^* . Then*

(i) [Sufficiency] *The Karush-Kuhn-Tucker condition:*

There exist nonnegative Lagrange multipliers $\lambda_j^, j = 1, \dots, m$, such that*

$$\lambda_j^* g_j(x^*) = 0, j = 1, \dots, m \quad [\text{complementary slackness}] \quad (\text{D.2.14})$$

and

$$\nabla f(x^*) + \sum_{j=1}^m \lambda_j^* \nabla g_j(x^*) \in N_X(x^*) \quad (\text{D.2.15})$$

$$(\text{that is, } (x - x^*)^T \nabla f(x^*) + \sum_{j=1}^m \lambda_j^* \nabla g_j(x^*) \geq 0 \text{ for all } x \in X)$$

is sufficient for x^ to be optimal solution to (IC).*

(ii) [Necessity and sufficiency] *If, in addition to the premise, the “restricted Slater assumption” holds, that is, there exists $\bar{x} \in X$ such that at \bar{x} the nonlinear g_j are strictly negative, and linear g_j are nonpositive, then the Karush-Kuhn-Tucker condition from (i) is necessary and sufficient for x^* to be optimal solution to (IC).*

Proof. (i) is readily given by Theorem D.2.3.(ii); indeed, it is immediately seen that under the premise of Theorem D.2.5 the Karush-Kuhn-Tucker condition is sufficient for x^*, λ^* to be a saddle point of the Lagrange function.

(ii) is contained in the proof of Theorem D.2.4. ■

Note that the optimality conditions stated in Theorem C.5.2 and Proposition C.5.1 are particular cases of the above Theorem corresponding to $m = 0$.

D.3 Duality in Linear and Convex Quadratic Programming

The fundamental role of the Lagrange function and Lagrange Duality in Optimization is clear already from the Optimality Conditions given by Theorem D.2.3, but this role is not restricted by this Theorem only. There are several cases when we can explicitly write down the Lagrange dual, and whenever it is the case, we get a pair of explicitly formulated and closely related to each other optimization programs – the *primal-dual pair*; analyzing the problems simultaneously, we get more information about their properties (and get a possibility to solve the problems numerically in a more efficient way) than it is possible when we restrict ourselves with only one problem of the pair. The detailed investigation of Duality in “well-structured” Convex Programming – in the cases when we can explicitly write down both the primal and the dual problems – goes beyond the scope of our course (mainly because the Lagrange duality is not the best possible approach here; the best approach is given by the *Fenchel Duality* – something similar, but not identical). There are, however, simple cases when already the Lagrange duality is quite appropriate. Let us look at two of these particular cases.

D.3.1 Linear Programming Duality

Let us start with some general observation. Note that the Karush-Kuhn-Tucker condition under the assumption of the Theorem ((IC) is convex, x^* is an interior point of X , f, g_1, \dots, g_m are

differentiable at x^*) is exactly the condition that $(x^*, \lambda^* = (\lambda_1^*, \dots, \lambda_m^*))$ is a saddle point of the Lagrange function

$$L(x, \lambda) = f(x) + \sum_{j=1}^m \lambda_j g_j(x) : \quad (\text{D.3.1})$$

(D.2.14) says that $L(x^*, \lambda)$ attains its maximum in $\lambda \geq 0$, and (D.2.15) says that $L(x, \lambda^*)$ attains its minimum in x at $x = x^*$.

Now consider the particular case of (IC) where $X = \mathbf{R}^n$ is the entire space, the objective f is convex and everywhere differentiable and the constraints g_1, \dots, g_m are linear. For this case, Theorem D.2.5 says to us that the KKT (Karush-Kuhn-Tucker) condition is necessary and sufficient for optimality of x^* ; as we just have explained, this is the same as to say that the necessary and sufficient condition of optimality for x^* is that x^* along with certain $\lambda^* \geq 0$ form a saddle point of the Lagrange function. Combining these observations with Proposition D.2.2, we get the following simple result:

Proposition D.3.1 *Let (IC) be a convex program with $X = \mathbf{R}^n$, everywhere differentiable objective f and linear constraints g_1, \dots, g_m . Then x^* is optimal solution to (IC) if and only if there exists $\lambda^* \geq 0$ such that (x^*, λ^*) is a saddle point of the Lagrange function (D.3.1) (regarded as a function of $x \in \mathbf{R}^n$ and $\lambda \geq 0$). In particular, (IC) is solvable if and only if L has saddle points, and if it is the case, then both (IC) and its Lagrange dual*

$$(\text{IC}^*) : \max_{\lambda} \{\underline{L}(\lambda) : \lambda \geq 0\}$$

are solvable with equal optimal values.

Let us look what this proposition says in the Linear Programming case, i.e., when (IC) is the program

$$(P) \quad \min_x \left\{ f(x) = c^T x : g_j(x) \equiv b_j - a_j^T x \leq 0, j = 1, \dots, m \right\}.$$

In order to get the Lagrange dual, we should form the Lagrange function

$$L(x, \lambda) = f(x) + \sum_{j=1}^m \lambda_j g_j(x) = [c - \sum_{j=1}^m \lambda_j a_j]^T x + \sum_{j=1}^m \lambda_j b_j$$

of (IC) and to minimize it in $x \in \mathbf{R}^n$; this will give us the dual objective. In our case the minimization in x is immediate: the minimal value is $-\infty$, if $c - \sum_{j=1}^m \lambda_j a_j \neq 0$, and is $\sum_{j=1}^m \lambda_j b_j$ otherwise. We see that the Lagrange dual is

$$(D) \quad \max_{\lambda} \left\{ b^T \lambda : \sum_{j=1}^m \lambda_j a_j = c, \lambda \geq 0 \right\}.$$

The problem we get is the usual LP dual to (P), and Proposition D.3.1 is one of the equivalent forms of the Linear Programming Duality Theorem which we already know.

D.3.2 Quadratic Programming Duality

Now consider the case when the original problem is linearly constrained convex quadratic program

$$(P) \quad \min_x \left\{ f(x) = \frac{1}{2} x^T D x + c^T x : g_j(x) \equiv b_j - a_j^T x \leq 0, j = 1, \dots, m \right\},$$

where the objective is a strictly convex quadratic form, so that $D = D^T$ is positive definite matrix: $x^T D x > 0$ whenever $x \neq 0$. It is convenient to rewrite the constraints in the vector-matrix form

$$g(x) = b - Ax \leq 0, \quad b = \begin{pmatrix} b_1 \\ \dots \\ b_m \end{pmatrix}, \quad A = \begin{pmatrix} a_1^T \\ \dots \\ a_m^T \end{pmatrix}.$$

In order to form the Lagrange dual to (P) program, we write down the Lagrange function

$$\begin{aligned} L(x, \lambda) &= f(x) + \sum_{j=1}^m \lambda_j g_j(x) \\ &= c^T x + \lambda^T (b - Ax) + \frac{1}{2} x^T D x \\ &= \frac{1}{2} x^T D x - [A^T \lambda - c]^T x + b^T \lambda \end{aligned}$$

and minimize it in x . Since the function is convex and differentiable in x , the minimum, if exists, is given by the Fermat rule

$$\nabla_x L(x, \lambda) = 0,$$

which in our situation becomes

$$Dx = [A^T \lambda - c].$$

Since D is positive definite, it is nonsingular, so that the Fermat equation has a unique solution which is the minimizer of $L(\cdot, \lambda)$; this solution is

$$x = D^{-1}[A^T \lambda - c].$$

Substituting the value of x into the expression for the Lagrange function, we get the dual objective:

$$\underline{L}(\lambda) = -\frac{1}{2}[A^T \lambda - c]^T D^{-1}[A^T \lambda - c] + b^T \lambda,$$

and the dual problem is to maximize this objective over the nonnegative orthant. Usually people rewrite this dual problem equivalently by introducing additional variables

$$t = -D^{-1}[A^T \lambda - c] \quad [[A^T \lambda - c]^T D^{-1}[A^T \lambda - c] = t^T D t];$$

with this substitution, the dual problem becomes

$$(D) \quad \max_{\lambda, t} \left\{ -\frac{1}{2} t^T D t + b^T \lambda : A^T \lambda + D t = c, \lambda \geq 0 \right\}.$$

We see that the dual problem also turns out to be linearly constrained convex quadratic program.

Note also that in the case in question feasible problem (P) automatically is solvable⁴⁾

With this observation, we get from Proposition D.3.1 the following

Theorem D.3.1 [Duality Theorem in Quadratic Programming]

Let (P) be feasible quadratic program with positive definite symmetric matrix D in the objective. Then both (P) and (D) are solvable, and the optimal values in the problems are equal to each other.

⁴⁾ since its objective, due to positive definiteness of D , goes to infinity as $|x| \rightarrow \infty$, and due to the following general fact:

Let (IC) be a feasible program with closed domain X , continuous on X objective and constraints and such that $f(x) \rightarrow \infty$ as $x \in X$ “goes to infinity” (i.e., $|x| \rightarrow \infty$). Then (IC) is solvable.

You are welcome to prove this simple statement (it is among the exercises accompanying the Lecture)

The pair $(x; (\lambda, t))$ of feasible solutions to the problems is comprised of the optimal solutions to them

(i) if and only if the primal objective at x is equal to the dual objective at (λ, t) [“zero duality gap” optimality condition]

same as

(ii) if and only if

$$\lambda_i(Ax - b)_i = 0, i = 1, \dots, m, \text{ and } t = -x. \quad (\text{D.3.2})$$

Proof. (i): we know from Proposition D.3.1 that the optimal value in minimization problem (P) is equal to the optimal value in the maximization problem (D) . It follows that the value of the primal objective at any primal feasible solution is \geq the value of the dual objective at any dual feasible solution, and equality is possible if and only if these values coincide with the optimal values in the problems, as claimed in (i).

(ii): Let us compute the difference Δ between the values of the primal objective at primal feasible solution x and the dual objective at dual feasible solution (λ, t) :

$$\begin{aligned} \Delta &= c^T x + \frac{1}{2} x^T D x - [b^T \lambda - \frac{1}{2} t^T D t] \\ &= [A^T \lambda + D t]^T x + \frac{1}{2} x^T D x + \frac{1}{2} t^T D t - b^T \lambda \\ &\quad [\text{since } A^T \lambda + D t = c] \\ &= \lambda^T [Ax - b] + \frac{1}{2} [x + t]^T D [x + t] \end{aligned}$$

Since $Ax - b \geq 0$ and $\lambda \geq 0$ due to primal feasibility of x and dual feasibility of (λ, t) , both terms in the resulting expression for Δ are nonnegative. Thus, $\Delta = 0$ (which, by (i). is equivalent to optimality of x for (P) and optimality of (λ, t) for (D)) if and only if $\sum_{j=1}^m \lambda_j (Ax - b)_j = 0$ and $(x + t)^T D (x + t) = 0$. The first of these equalities, due to $\lambda \geq 0$ and $Ax \geq b$, is equivalent to $\lambda_j (Ax - b)_j = 0, j = 1, \dots, m$; the second, due to positive definiteness of D , is equivalent to $x + t = 0$. ■

D.4 Saddle Points

D.4.1 Definition and Game Theory interpretation

When speaking about the “saddle point” formulation of optimality conditions in Convex Programming, we touched a very interesting in its own right topic of Saddle Points. This notion is related to the situation as follows. Let $X \subset \mathbf{R}^n$ and $\Lambda \in \mathbf{R}^m$ be two nonempty sets, and let

$$L(x, \lambda) : X \times \Lambda \rightarrow \mathbf{R}$$

be a real-valued function of $x \in X$ and $\lambda \in \Lambda$. We say that a point $(x^*, \lambda^*) \in X \times \Lambda$ is a *saddle point* of L on $X \times \Lambda$, if L attains in this point its maximum in $\lambda \in \Lambda$ and attains at the point its minimum in $x \in X$:

$$L(x, \lambda^*) \geq L(x^*, \lambda^*) \geq L(x^*, \lambda) \quad \forall (x, \lambda) \in X \times \Lambda. \quad (\text{D.4.1})$$

The notion of a saddle point admits natural interpretation in *game terms*. Consider what is called a *two person zero sum game* where player I chooses $x \in X$ and player II chooses $\lambda \in \Lambda$; after the players have chosen their decisions, player I pays to player II the sum $L(x, \lambda)$. Of course, I is interested to minimize his payment, while II is interested to maximize his income.

What is the natural notion of the equilibrium in such a game – what are the choices (x, λ) of the players I and II such that every one of the players is not interested to vary his choice independently on whether he knows the choice of his opponent? It is immediately seen that the equilibria are exactly the saddle points of the cost function L . Indeed, if (x^*, λ^*) is such a point, than the player I is not interested to pass from x to another choice, given that II keeps his choice λ fixed: the first inequality in (D.4.1) shows that such a choice cannot decrease the payment of I. Similarly, player II is not interested to choose something different from λ^* , given that I keeps his choice x^* – such an action cannot increase the income of II. On the other hand, if (x^*, λ^*) is not a saddle point, then either the player I can decrease his payment passing from x^* to another choice, given that II keeps his choice at λ^* – this is the case when the first inequality in (D.4.1) is violated, or similarly for the player II; thus, equilibria are exactly the saddle points.

The game interpretation of the notion of a saddle point motivates deep insight into the structure of the set of saddle points. Consider the following two situations:

(A) player I makes his choice first, and player II makes his choice already knowing the choice of I;

(B) vice versa, player II chooses first, and I makes his choice already knowing the choice of II.

In the case (A) the reasoning of I is: If I choose some x , then II of course will choose λ which maximizes, for my x , my payment $L(x, \lambda)$, so that I shall pay the sum

$$\bar{L}(x) = \sup_{\lambda \in \Lambda} L(x, \lambda);$$

Consequently, my policy should be to choose x which minimizes my *loss function* L , i.e., the one which solves the optimization problem

$$(I) \quad \min_{x \in X} \bar{L}(x);$$

with this policy my anticipated payment will be

$$\inf_{x \in X} \bar{L}(x) = \inf_{x \in X} \sup_{\lambda \in \Lambda} L(x, \lambda).$$

In the case (B), similar reasoning of II enforces him to choose λ maximizing his *profit function*

$$\underline{L}(\lambda) = \inf_{x \in X} L(x, \lambda),$$

i.e., the one which solves the optimization problem

$$(II) \quad \max_{\lambda \in \Lambda} \underline{L}(\lambda);$$

with this policy, the anticipated profit of II is

$$\sup_{\lambda \in \Lambda} \underline{L}(\lambda) = \sup_{\lambda \in \Lambda} \inf_{x \in X} L(x, \lambda).$$

Note that these two reasonings relate to two *different games*: the one with priority of II (when making his decision, II already knows the choice of I), and the one with similar priority of I. Therefore we should not, generally speaking, expect that the anticipated loss of I in (A) is equal to the anticipated profit of II in (B). What can be guessed is that the anticipated loss of I in (B) is *less than or equal* to the anticipated profit of II in (A), since the conditions of the game

(B) are better for I than those of (A). Thus, we may guess that independently of the structure of the function $L(x, \lambda)$, there is the inequality

$$\sup_{\lambda \in \Lambda} \inf_{x \in X} L(x, \lambda) \leq \inf_{x \in X} \sup_{\lambda \in \Lambda} L(x, \lambda). \quad (\text{D.4.2})$$

This inequality indeed is true; which is seen from the following reasoning:

$$\begin{aligned} \forall y \in X : \inf_{x \in X} L(x, \lambda) &\leq L(y, \lambda) \Rightarrow \\ \forall y \in X : \sup_{\lambda \in \Lambda} \inf_{x \in X} L(x, \lambda) &\leq \sup_{\lambda \in \Lambda} L(y, \lambda) \equiv \underline{L}(y); \end{aligned}$$

consequently, the quantity $\sup_{\lambda \in \Lambda} \inf_{x \in X} L(x, \lambda)$ is a lower bound for the function $\underline{L}(y)$, $y \in X$, and is therefore a lower bound for the infimum of the latter function over $y \in X$, i.e., is a lower bound for $\inf_{y \in X} \sup_{\lambda \in \Lambda} L(y, \lambda)$.

Now let us look what happens when the game in question has a saddle point (x^*, λ^*) , so that

$$L(x, \lambda^*) \geq L(x^*, \lambda^*) \geq L(x^*, \lambda) \quad \forall (x, \lambda) \in X \times \Lambda. \quad (\text{D.4.3})$$

I claim that if it is the case, then

(*) x^* is an optimal solution to (I), λ^* is an optimal solution to (II) and the optimal values in these two optimization problems are equal to each other (and are equal to the quantity $L(x^*, \lambda^*)$).

Indeed, from (D.4.3) it follows that

$$\underline{L}(\lambda^*) \geq L(x^*, \lambda^*) \geq \overline{L}(x^*),$$

whence, of course,

$$\sup_{\lambda \in \Lambda} \underline{L}(\lambda) \geq \underline{L}(\lambda^*) \geq L(x^*, \lambda^*) \geq \overline{L}(x^*) \geq \inf_{x \in X} \overline{L}(x).$$

the very first quantity in the latter chain is \leq the very last quantity by (D.4.2), which is possible if and only if all the inequalities in the chain are equalities, which is exactly what is said by (A) and (B).

Thus, if (x^*, λ^*) is a saddle point of L , then (*) takes place. We are about to demonstrate that the inverse also is true:

Theorem D.4.1 [Structure of the saddle point set] *Let $L : X \times Y \rightarrow \mathbf{R}$ be a function. The set of saddle points of the function is nonempty if and only if the related optimization problems (I) and (II) are solvable and the optimal values in the problems are equal to each other. If it is the case, then the saddle points of L are exactly all pairs (x^*, λ^*) with x^* being an optimal solution to (I) and λ^* being an optimal solution to (II), and the value of the cost function $L(\cdot, \cdot)$ at every one of these points is equal to the common optimal value in (I) and (II).*

Proof. We already have established “half” of the theorem: if there are saddle points of L , then their components are optimal solutions to (I), respectively, (II), and the optimal values in these two problems are equal to each other and to the value of L at the saddle point in question. To complete the proof, we should demonstrate that if x^* is an optimal solution to (I), λ^* is an

optimal solution to (II) and the optimal values in the problems are equal to each other, then (x^*, λ^*) is a saddle point of L . This is immediate: we have

$$\begin{aligned} L(x, \lambda^*) &\geq \underline{L}(\lambda^*) && [\text{definition of } \underline{L}] \\ &= \overline{L}(x^*) && [\text{by assumption}] \\ &\geq L(x^*, \lambda) && [\text{definition of } \overline{L}] \end{aligned}$$

whence

$$L(x, \lambda^*) \geq L(x^*, \lambda) \quad \forall x \in X, \lambda \in \Lambda;$$

substituting $\lambda = \lambda^*$ in the right hand side of this inequality, we get $L(x, \lambda^*) \geq L(x^*, \lambda^*)$, and substituting $x = x^*$ in the right hand side of our inequality, we get $L(x^*, \lambda^*) \geq L(x^*, \lambda)$; thus, (x^*, λ^*) indeed is a saddle point of L . ■

D.4.2 Existence of Saddle Points

It is easily seen that a "quite respectable" cost function may have no saddle points, e.g., the function $L(x, \lambda) = (x - \lambda)^2$ on the unit square $[0, 1] \times [0, 1]$. Indeed, here

$$\underline{L}(x) = \sup_{\lambda \in [0,1]} (x - \lambda)^2 = \max\{x^2, (1-x)^2\},$$

$$\overline{L}(\lambda) = \inf_{x \in [0,1]} (x - \lambda)^2 = 0, \quad \lambda \in [0, 1],$$

so that the optimal value in (I) is $\frac{1}{4}$, and the optimal value in (II) is 0; according to Theorem D.4.1 it means that L has no saddle points.

On the other hand, there are generic cases when L has a saddle point, e.g., when

$$L(x, \lambda) = f(x) + \sum_{i=1}^m \lambda_i g_i(x) : X \times \mathbf{R}_+^m \rightarrow \mathbf{R}$$

is the Lagrange function of a solvable convex program satisfying the Slater condition. Note that in this case L is convex in x for every $\lambda \in \Lambda \equiv \mathbf{R}_+^m$ and is linear (and therefore concave) in λ for every fixed X . As we shall see in a while, these are the structural properties of L which take upon themselves the "main responsibility" for the fact that in the case in question the saddle points exist. Namely, there exists the following

Theorem D.4.2 [Existence of saddle points of a convex-concave function (Sion-Kakutani, Bounded case)] *Let X and Λ be convex compact sets in \mathbf{R}^n and \mathbf{R}^m , respectively, and let*

$$L(x, \lambda) : X \times \Lambda \rightarrow \mathbf{R}$$

be a continuous function which is convex in $x \in X$ for every fixed $\lambda \in \Lambda$ and is concave in $\lambda \in \Lambda$ for every fixed $x \in X$. Then L has saddle points on $X \times \Lambda$.

Proof. According to Theorem D.4.1, we should prove that

- (i) Optimization problems (I) and (II) are solvable
- (ii) the optimal values in (I) and (II) are equal to each other.

(i) is valid independently of convexity-concavity of L and is given by the following routine reasoning from the Analysis:

Since X and Λ are compact sets and L is continuous on $X \times \Lambda$, due to the well-known Analysis theorem L is uniformly continuous on $X \times \Lambda$: for every $\epsilon > 0$ there exists $\delta(\epsilon) > 0$ such that

$$|x - x'| + |\lambda - \lambda'| \leq \delta(\epsilon) \Rightarrow |L(x, \lambda) - L(x', \lambda')| \leq \epsilon \quad ^5) \quad (\text{D.4.4})$$

In particular,

$$|x - x'| \leq \delta(\epsilon) \Rightarrow |L(x, \lambda) - L(x', \lambda)| \leq \epsilon,$$

whence, of course, also

$$|x - x'| \leq \delta(\epsilon) \Rightarrow |\bar{L}(x) - \bar{L}(x')| \leq \epsilon,$$

so that the function \bar{L} is continuous on X . Similarly, \underline{L} is continuous on Λ . Taking in account that X and Λ are compact sets, we conclude that the problems (I) and (II) are solvable.

(ii) is the essence of the matter; here, of course, the entire construction heavily exploits convexity-concavity of L .

0⁰. To prove (ii), we first establish the following statement, which is important by its own right:

Lemma D.4.1 [Minmax Lemma] *Let X be a convex compact set and f_0, \dots, f_N be a collection of $N + 1$ convex and continuous functions on X . Then the minmax*

$$\min_{x \in X} \max_{i=0, \dots, N} f_i(x) \quad (\text{D.4.5})$$

of the collection is equal to the minimum in $x \in X$ of certain convex combination of the functions: there exist nonnegative μ_i , $i = 0, \dots, N$, with unit sum such that

$$\min_{x \in X} \max_{i=0, \dots, N} f_i(x) = \min_{x \in X} \sum_{i=0}^N \mu_i f_i(x)$$

Remark D.4.1 Minimum of every convex combination of a collection of arbitrary functions is \leq the minmax of the collection; this evident fact can be also obtained from (D.4.2) as applied to the function

$$M(x, \mu) = \sum_{i=0}^N \mu_i f_i(x)$$

on the direct product of X and the standard simplex

$$\Delta = \{\mu \in \mathbf{R}^{N+1} \mid \mu \geq 0, \sum_i \mu_i = 1\}.$$

The Minmax Lemma says that if f_i are convex and continuous on a convex compact set X , then the indicated inequality is in fact equality; you can easily verify that this is nothing but the claim that the function M possesses a saddle point. Thus, the Minmax Lemma is in fact a particular case of the Sion-Kakutani Theorem; we are about to give a direct proof of this particular case of the Theorem and then to derive the general case from this particular one.

⁵⁾ for those not too familiar with Analysis, I wish to stress the difference between the usual continuity and the uniform continuity: continuity of L means that given $\epsilon > 0$ and a point (x, λ) , it is possible to choose $\delta > 0$ such that (D.4.4) is valid; the corresponding δ may depend on (x, λ) , not only on ϵ . Uniform continuity means that this positive δ may be chosen as a function of ϵ only. The fact that a continuous on a compact set function automatically is uniformly continuous on the set is one of the most useful features of compact sets

Proof of the Minmax Lemma. Consider the optimization program

$$(S) \quad \min_{t,x} \{t : f_0(x) - t \leq 0, f_1(x) - t \leq 0, \dots, f_N(x) - t \leq 0, x \in X\}.$$

This clearly is a convex program with the optimal value

$$t^* = \min_{x \in X} \max_{i=0, \dots, N} f_i(x)$$

(note that (t, x) is feasible solution for (S) if and only if $x \in X$ and $t \geq \max_{i=0, \dots, N} f_i(x)$). The problem clearly satisfies the Slater condition and is solvable (since X is compact set and f_i , $i = 0, \dots, N$, are continuous on X ; therefore their maximum also is continuous on X and thus attains its minimum on the compact set X). Let (t^*, x^*) be an optimal solution to the problem. According to Theorem D.2.3, there exists $\lambda^* \geq 0$ such that $((t^*, x^*), \lambda^*)$ is a saddle point of the corresponding Lagrange function

$$L(t, x; \lambda) = t + \sum_{i=0}^N \lambda_i (f_i(x) - t) = t(1 - \sum_{i=0}^N \lambda_i) + \sum_{i=0}^N \lambda_i f_i(x),$$

and the value of this function at $((t^*, x^*), \lambda^*)$ is equal to the optimal value in (S) , i.e., to t^* .

Now, since $L(t, x; \lambda^*)$ attains its minimum in (t, x) over the set $\{t \in \mathbf{R}, x \in X\}$ at (t^*, x^*) , we should have

$$\sum_{i=0}^N \lambda_i^* = 1$$

(otherwise the minimum of L in (t, x) would be $-\infty$). Thus,

$$[\min_{x \in X} \max_{i=0, \dots, N} f_i(x) =] \quad t^* = \min_{t \in \mathbf{R}, x \in X} \left[t \times 0 + \sum_{i=0}^N \lambda_i^* f_i(x) \right],$$

so that

$$\min_{x \in X} \max_{i=0, \dots, N} f_i(x) = \min_{x \in X} \sum_{i=0}^N \lambda_i^* f_i(x)$$

with some $\lambda_i^* \geq 0$, $\sum_{i=0}^N \lambda_i^* = 1$, as claimed. ■

From the Minmax Lemma to the Sion-Kakutani Theorem. We should prove that the optimal values in (I) and (II) (which, by (i), are well defined reals) are equal to each other, i.e., that

$$\inf_{x \in X} \sup_{\lambda \in \Lambda} L(x, \lambda) = \sup_{\lambda \in \Lambda} \inf_{x \in X} L(x, \lambda).$$

We know from (D.4.4) that the first of these two quantities is greater than or equal to the second, so that all we need is to prove the inverse inequality. For me it is convenient to assume that the right quantity (the optimal value in (II)) is 0, which, of course, does not restrict generality; and all we need to prove is that the left quantity – the optimal value in (I) – cannot be positive.

¹⁰. What does it mean that the optimal value in (II) is zero? When it is zero, then the function $\underline{L}(\lambda)$ is nonpositive for every λ , or, which is the same, the convex continuous function of

$x \in X$ – the function $L(x, \lambda)$ – has nonpositive minimal value over $x \in X$. Since X is compact, this minimal value is achieved, so that the set

$$X(\lambda) = \{x \in X \mid L(x, \lambda) \leq 0\}$$

is nonempty; and since X is convex and L is convex in $x \in X$, the set $X(\lambda)$ is convex (as a level set of a convex function, Proposition C.1.4). Note also that the set is closed (since X is closed and $L(x, \lambda)$ is continuous in $x \in X$).

2⁰. Thus, if the optimal value in (II) is zero, then the set $X(\lambda)$ is a nonempty convex compact set for every $\lambda \in \Lambda$. And what does it mean that the optimal value in (I) is nonpositive? It means exactly that there is a point $x \in X$ where the function \bar{L} is nonpositive, i.e., the point $x \in X$ where $L(x, \lambda) \leq 0$ for all $\lambda \in \Lambda$. In other words, to prove that the optimal value in (I) is nonpositive is the same as to prove that *the sets $X(\lambda)$, $\lambda \in \Lambda$, have a point in common*.

3⁰. With the above observations we see that the situation is as follows: we are given a family of closed nonempty convex subsets $X(\lambda)$, $\lambda \in \Lambda$, of a compact set X , and we should prove that these sets have a point in common. To this end, in turn, it suffices to prove that every *finite* number of sets from our family have a point in common (to justify this claim, I can refer to the Helly Theorem II, which gives us much stronger result: to prove that all $X(\lambda)$ have a point in common, it suffices to prove that every $(n+1)$ sets of this family, n being the affine dimension of X , have a point in common). Let $X(\lambda_0), \dots, X(\lambda_N)$ be $N+1$ sets from our family; we should prove that the sets have a point in common. In other words, let

$$f_i(x) = L(x, \lambda_i), \quad i = 0, \dots, N;$$

all we should prove is that there exists a point x where all our functions are nonpositive, or, which is the same, that the minmax of our collection of functions – the quantity

$$\alpha \equiv \min_{x \in X} \max_{i=1, \dots, N} f_i(x)$$

– is nonpositive.

The proof of the inequality $\alpha \leq 0$ is as follows. According to the Minmax Lemma (which can be applied in our situation – since L is convex and continuous in x , all f_i are convex and continuous, and X is compact), α is the minimum in $x \in X$ of certain convex combination $\phi(x) = \sum_{i=0}^N \nu_i f_i(x)$ of the functions $f_i(x)$. We have

$$\phi(x) = \sum_{i=0}^N \nu_i f_i(x) \equiv \sum_{i=0}^N \nu_i L(x, \lambda_i) \leq L(x, \sum_{i=0}^N \nu_i \lambda_i)$$

(the last inequality follows from concavity of L in λ ; this is the only – and crucial – point where we use this assumption). We see that $\phi(\cdot)$ is majorated by $L(\cdot, \lambda)$ for a properly chosen λ ; it follows that the minimum of ϕ in $x \in X$ – and we already know that this minimum is exactly α – is nonpositive (recall that the minimum of L in x is nonpositive for every λ). ■

The next theorem lifts the assumption of boundedness of X and Λ in Theorem D.4.2 – now only one of these sets should be bounded – at the price of some weakening of the conclusion.

Theorem D.4.3 [Swapping min and max in convex-concave saddle point problem (Sion-Kakutani)] *Let X and Λ be convex sets in \mathbf{R}^n and \mathbf{R}^m , respectively, with X being compact, and let*

$$L(x, \lambda) : X \times \Lambda \rightarrow \mathbf{R}$$

be a continuous function which is convex in $x \in X$ for every fixed $\lambda \in \Lambda$ and is concave in $\lambda \in \Lambda$ for every fixed $x \in X$. Then

$$\inf_{x \in X} \sup_{\lambda \in \Lambda} L(x, \lambda) = \sup_{\lambda \in \Lambda} \inf_{x \in X} L(x, \lambda). \quad (\text{D.4.6})$$

Proof. By general theory, in (D.4.6) the left hand side is \geq the right hand side, so that there is nothing to prove when the right hand side is $+\infty$. Assume that this is not the case. Since X is compact and L is continuous in $x \in X$, $\inf_{x \in X} L(x, \lambda) > -\infty$ for every $\lambda \in \Lambda$, so that the left hand side in (D.4.6) cannot be $-\infty$; since it is not $+\infty$ as well, it is a real, and by shift, we can assume w.l.o.g. that this real is 0:

$$\sup_{\lambda \in \Lambda} \inf_{x \in X} L(x, \lambda) = 0.$$

All we need to prove now is that the left hand side in (D.4.6) is nonpositive. Assume, on the contrary, that it is positive and thus is $> c$ with some $c > 0$. Then for every $x \in X$ there exists $\lambda_x \in \Lambda$ such that $L(x, \lambda_x) > c$. By continuity, there exists a neighborhood V_x of x on X such that $L(x', \lambda_x) \geq c$ for all $x' \in V_x$. Since X is compact, we can find finitely many points x_1, \dots, x_n in X such that the union, over $1 \leq i \leq n$, of V_{x_i} is exactly X , implying that $\max_{1 \leq i \leq n} L(x, \lambda_{x_i}) \geq c$ for every $x \in X$. Now let $\bar{\Lambda}$ be the convex hull of $\{\lambda_{x_1}, \dots, \lambda_{x_n}\}$, so that $\max_{\lambda \in \bar{\Lambda}} L(x, \lambda) \geq c$ for every $x \in X$. Applying to L and the convex compact sets $X, \bar{\Lambda}$ Theorem D.4.2, we get the equality in the following chain:

$$c \leq \min_{x \in X} \max_{\lambda \in \bar{\Lambda}} L(x, \lambda) = \max_{\lambda \in \bar{\Lambda}} \min_{x \in X} L(x, \lambda) \leq \sup_{\lambda \in \Lambda} \min_{x \in X} L(x, \lambda) = 0,$$

which is a desired contradiction (recall that $c > 0$). ■

Slightly strengthening the premise of Theorem D.4.3, we can replace (D.4.6) with existence of a saddle point:

Theorem D.4.4 [Existence of saddle point in convex-concave saddle point problem (Sion-Kakutani, Semi-Bounded case)] *Let X and Λ be closed convex sets in \mathbf{R}^n and \mathbf{R}^m , respectively, with X being compact, and let*

$$L(x, \lambda) : X \times \Lambda \rightarrow \mathbf{R}$$

be a continuous function which is convex in $x \in X$ for every fixed $\lambda \in \Lambda$ and is concave in $\lambda \in \Lambda$ for every fixed $x \in X$. Assume that for every $a \in \mathbf{R}$, there exist a collection $x_1^a, \dots, x_{n_a}^a \in X$ such that the set

$$\{\lambda \in \Lambda : L(x_i^a, \lambda) \geq a \mid 1 \leq i \leq n_a\}$$

is bounded⁶. Then L has saddle points on $X \times \Lambda$.

Proof. Since X is compact and L is continuous, the function $\underline{L}(\lambda) = \min_{x \in X} L(x, \lambda)$ is real-valued and continuous on Λ . Further, for every $a \in \mathbf{R}$, the set $\{\lambda \in \Lambda : \underline{L}(\lambda) \geq a\}$ clearly is contained in the set $\{\lambda : L(x_i^a, \lambda) \geq a, 1 \leq i \leq n_a\}$ and thus is bounded. Thus, $\underline{L}(\lambda)$ is a continuous function on a closed set Λ , and the level sets $\{\lambda \in \Lambda : \underline{L}(\lambda) \geq a\}$ are bounded, implying that \underline{L} attains its maximum on Λ . Invoking Theorem D.4.3, it follows that $\inf_{x \in X} [\bar{L}(x) := \sup_{\lambda \in \Lambda} L(x, \lambda)]$ is finite, implying that the function $\bar{L}(\cdot)$ is not $+\infty$ identically in $x \in X$. Since L is continuous,

⁶this definitely is the case true when $L(\bar{x}, \lambda)$ is coercive in λ for some $\bar{x} \in X$, meaning that the sets $\{\lambda \in \Lambda : L(\bar{x}, \lambda) \geq a\}$ are bounded for every $a \in \mathbf{R}$, or, equivalently, whenever $\lambda_i \in \Lambda$ and $\|\lambda_i\|_2 \rightarrow \infty$ as $i \rightarrow \infty$, we have $L(\bar{x}, \lambda_i) \rightarrow -\infty$ as $i \rightarrow \infty$.

\bar{L} is lower semicontinuous. Thus, $\bar{L} : X \rightarrow \mathbf{R} \cup \{+\infty\}$ is a lower semicontinuous proper (i.e., not identically $+\infty$) function on X ; since X is compact, \bar{L} attains its minimum on X . Thus, both problems $\max_{\lambda \in \Lambda} \underline{L}(\lambda)$ and $\min_{x \in X} \bar{L}(x)$ are solvable, and the optimal values in the problem are equal by Theorem D.4.3. Invoking Theorem D.4.1, L has a saddle point. ■

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