

Introduction to Differential Manifolds

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Chapter 1

Manifolds and Vector Fields

1.1 Manifolds and Maps

The title of this course is “Introduction to Differential Manifolds,” which suggests that these *differential manifolds* (or sometimes *differentiable manifolds*), whatever they are, will probably be important. So what is a differential manifold? The name should suggest the answer: they are spaces in which we know what differentiation is supposed to mean. I actually prefer the term *smooth manifold*, so that is what I will use going forward, though this will often just get shortened to *manifold*.

In practice, the idea is to leverage the fact that we already (hopefully!) know how to do calculus in \mathbb{R}^n (this is exactly what MATH 261 is all about), and to translate those techniques to more general spaces. The key insight here is that differentiation is a local operation: to compute a derivative at a point (whether it’s a gradient, curl, divergence, directional derivative, whatever), you really only need to know what’s going on in a tiny open neighborhood of that point.

So to get a space on which we can compute derivatives, it’s enough to have a space which is “locally Euclidean” or “locally like \mathbb{R}^n ,” and this is what manifolds are. Roughly speaking, this means that around any point in a manifold you can find a small open set which looks just like an open set in some \mathbb{R}^n , and then we can do calculus on the manifold by translating in a neighborhood of a point to the corresponding set in \mathbb{R}^n , where we know what to do.

This is all to say that the point of defining manifolds in the way we are about to (which is extremely non-obvious and unintuitive!) is that these are precisely the spaces in which a suitable generalization of multivariable calculus makes sense.

So what does “locally like \mathbb{R}^n ” actually mean? Here’s a standard definition:

Definition 1.1.1. A *smooth manifold of dimension n* is a Hausdorff, second-countable topological space M together with a family of injective maps $\phi_\alpha : U_\alpha \rightarrow M$ from open sets $U_\alpha \subseteq \mathbb{R}^n$ so that:

- (i) $\bigcup_\alpha \phi_\alpha(U_\alpha) = M$ (that is, the images of the maps ϕ_α cover all of M);
- (ii) For any α, β so that $\phi_\alpha(U_\alpha) \cap \phi_\beta(U_\beta) = W \neq \emptyset$, the sets $\phi_\alpha^{-1}(W)$ and $\phi_\beta^{-1}(W)$ are open sets in \mathbb{R}^n and the maps $\phi_\beta^{-1} \circ \phi_\alpha$ and $\phi_\alpha^{-1} \circ \phi_\beta$ (when restricted to these open sets) are smooth.
- (iii) The family $\{(U_\alpha, \phi_\alpha)\}$ is maximal with respect to (i) and (ii).

The pairs (U_α, ϕ_α) are called *coordinate charts* and the (maximal) collection $\{(U_\alpha, \phi_\alpha)\}$ is called an *atlas*.

See Figure 1.1 for a visualization of (ii), showing the map $\phi_\beta^{-1} \circ \phi_\alpha$ from $\phi_\alpha^{-1}(W) \subseteq \mathbb{R}^n$ to $\phi_\beta^{-1}(W) \subseteq \mathbb{R}^n$.



Figure 1.1: Transition maps.

Remark 1.1.2. In (ii) above, “smooth” means “infinitely differentiable” or, in shorthand, C^∞ . What I’m calling *smooth manifolds* are sometimes also called *C^∞ manifolds*. More generally we can talk about C^α manifolds for any integer $\alpha \geq 0$, where we just modify (ii) to require the maps $\phi_\beta^{-1} \circ \phi_\alpha$ and $\phi_\alpha^{-1} \circ \phi_\beta$ to be C^α .¹ In the special case $\alpha = 0$, this is just a requirement that these maps be continuous, and C^0 manifolds are often called *topological manifolds*.

Example 1.1.3. The maximal family containing $(\mathbb{R}^n, \text{id})$ makes \mathbb{R}^n into a smooth manifold.

Of course, most interesting manifolds are not \mathbb{R}^n , but the idea of (i) from Definition 1.1.1 is that you can cover any manifold by a bunch of little open sets (namely, the $\phi_\alpha(U_\alpha)$) that are essentially identical to open sets in \mathbb{R}^n (namely, the U_α),² so you can essentially do any local calculation in \mathbb{R}^n . It’s also very important that the n is always the same here: if $m \neq n$, we’re not allowed to have some points with neighborhoods that look like \mathbb{R}^m and some other points whose neighborhoods look like \mathbb{R}^n .

If the idea is to use the coordinate charts to transport calculations from the manifold to \mathbb{R}^n , then a thing you should be very worried about is that, if a point lies in two different charts, then there are two different ways to do this and they might not be compatible. This is the point of (ii): whether you do your calculations in U_α or U_β , the two are related by a smooth map, so you can easily translate between the two calculations using the change-of-variables formula. Indeed, the usual English-language gloss of (ii) is that “transition functions are smooth.”

¹Recall that a continuous map is C^α if it has α continuous derivatives.

²In topological terms, U_α and $\phi_\alpha(U_\alpha)$ are homeomorphic.

Finally, (iii) is a technical condition that in practice is not important. The point of it is simply to ensure uniqueness: if you had a collection of coordinate charts satisfying (i) and (ii), and I took your collection and added some new charts while still satisfying (ii), it would be kind of silly to say that you and I were talking about different manifolds. Taking maximal families gives uniqueness since your collection of charts and my collection of charts live in the same maximal family.

However, it is certainly possible to have distinct maximal collections on the same space (if there is one that is in some sense standard, then any others are sometimes called “exotic smooth structures”). At least two Fields Medals have been awarded primarily for finding examples of exotic smooth structures: to John Milnor in 1962 (for finding exotic 7-spheres [22]; it turns out there are exactly 28 distinct smooth structures on S^7 [18]) and to Simon Donaldson in 1986 (for finding exotic \mathbb{R}^4 ’s [8, 10, 14]; it turns out there are uncountably many distinct smooth structures on \mathbb{R}^4 [26]). It remains an open problem called the *smooth 4-dimensional Poincaré conjecture* whether there are non-standard differentiable structures on S^4 .

Remark 1.1.4. We often mimic the \mathbb{R}^n notation and indicate the dimension of a manifold M with a superscript; i.e., M^n means that M is an n -dimensional manifold, not that we are taking the Cartesian product $M \times M \times \dots \times M$.

Example 1.1.5. S^n the unit sphere in \mathbb{R}^{n+1} is a manifold. Specifically, I claim that the maximal family containing $\{(R^n, \phi_N), (R^n, \phi_S)\}$ makes S^n into an n -dimensional manifold, where ϕ_N and ϕ_S are inverse stereographic projection from the north and south poles, respectively.

Specifically, with $\vec{x} = (x_1, \dots, x_n) \in \mathbb{R}^n$, define

$$\phi_N(\vec{x}) := \frac{1}{1 + \|\vec{x}\|^2} (2x_1, \dots, 2x_n, -1 + \|\vec{x}\|^2)$$

and

$$\phi_S(\vec{x}) := \frac{1}{1 + \|\vec{x}\|^2} (2x_1, \dots, 2x_n, 1 - \|\vec{x}\|^2).$$

Then ϕ_N is the map that sends $\vec{x} \in \mathbb{R}^n$ to the point on the sphere which lies on the line segment connecting $(x_1, \dots, x_n, 0) \in \mathbb{R}^{n+1}$ to the north pole $(0, \dots, 0, 1) \in S^n \subset \mathbb{R}^{n+1}$; see Figure 1.2.

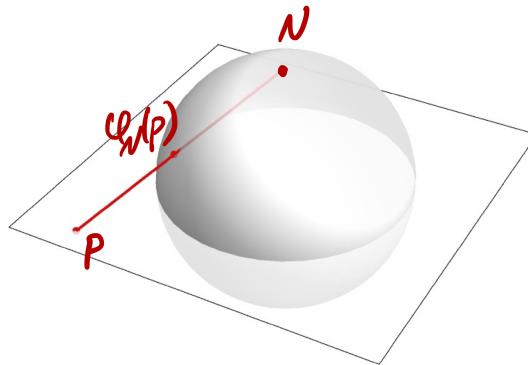


Figure 1.2: Inverse stereographic projection.

To prove the claim, we need to show that (i) and (ii) from Definition 1.1.1 are satisfied ((iii) is automatically satisfied, since we’re taking the maximal family containing $\{(\mathbb{R}^n, \phi_N), (\mathbb{R}^n, \phi_S)\}$).

If we let $N = (0, \dots, 0, 1)$ be the north pole and $S = (0, \dots, 0, -1)$ the south pole, then $\phi_N(\mathbb{R}^n) = S^n \setminus \{N\}$ and $\phi_S(\mathbb{R}^n) = S^n \setminus \{S\}$ and the union is all of S^n , so (i) is satisfied.

For (ii), observe that

$$W = \phi_N(\mathbb{R}^n) \cap \phi_S(\mathbb{R}^n) = S^n \setminus \{N, S\},$$

so

$$\phi_N^{-1}(W) = \mathbb{R}^n \setminus \{0\},$$

which is certainly open, and likewise for $\phi_S^{-1}(W) = \mathbb{R}^n \setminus \{0\}$. So we need to verify that $\phi_N^{-1} \circ \phi_S$ and $\phi_S^{-1} \circ \phi_N$ are smooth as functions on $\mathbb{R}^n \setminus \{0\}$.

The inverse of ϕ_N is stereographic projection

$$\phi_N^{-1}(\vec{y}) := \frac{1}{1 - y_{n+1}}(y_1, \dots, y_n)$$

(check this!) so we see that

$$(\phi_N^{-1} \circ \phi_S)(\vec{x}) = \frac{1}{\|\vec{x}\|^2} \vec{x}$$

is reflection through the unit sphere in \mathbb{R}^n , which is smooth away from the origin. And similarly for $\phi_S^{-1} \circ \phi_N$.

As already mentioned, the point of manifolds is that they are spaces in which we can do calculus, so we should be able to say what it means for a map between manifolds to be differentiable. Hopefully it's already starting to become clear what the strategy is: we can talk about differentiability at a point, and then both the point in the domain and the point it maps to in the range lie in coordinate charts that are like open sets in Euclidean spaces. So then locally our map just looks like a map between Euclidean spaces, where we already know what it means for a map to be differentiable.

Definition 1.1.6. Let M^m and N^n be manifolds. A continuous map $f: M \rightarrow N$ is *differentiable* at $p \in M$ if, given a coordinate chart $\psi: V \subseteq \mathbb{R}^n \rightarrow N$ containing $f(p)$, there exists a coordinate chart $\phi: U \subseteq \mathbb{R}^m \rightarrow M$ containing p so that $f(\phi(U)) \subseteq \psi(V)$ and

$$\psi^{-1} \circ f \circ \phi: U \subseteq \mathbb{R}^m \rightarrow \mathbb{R}^n$$

is differentiable at $\phi^{-1}(p)$ (see Figure 1.3). The map f is differentiable on an open set in M if it is differentiable at every point in that set.

Example 1.1.7. Consider the antipodal map $\alpha: S^n \rightarrow S^n$ given by $\alpha(\vec{y}) = -\vec{y}$. Then, so long as \vec{y} is not the south pole, $\alpha(\vec{y}) = -\vec{y} \in \phi_N(\mathbb{R}^n)$ so we can take $\psi = \phi_N$ and $V = \mathbb{R}^n$. Moreover, $\vec{y} \in \phi_S(\mathbb{R}^n)$ so we can $\phi = \phi_S$ and $U = \mathbb{R}^n$, since $\alpha(\phi_S(\mathbb{R}^n)) = S^n \setminus \{N\} = \phi_N(\mathbb{R}^n)$. Then a straightforward calculation shows that

$$(\phi_N^{-1} \circ \alpha \circ \phi_S)(\vec{x}) = -\vec{x},$$

which is definitely differentiable everywhere (as a map $\mathbb{R}^n \rightarrow \mathbb{R}^n$).

Of course, if $\vec{y} = S$, we can swap the roles of ϕ_N and ϕ_S in the above, and we conclude that α is differentiable everywhere on S^n .

While we've given the definition of a manifold and of a differentiable map in this section, we generally try to use them directly as little as possible. They are hard to handle and fairly unintuitive, so we will quickly be looking for alternative ways of characterizing manifolds and differentiable maps.

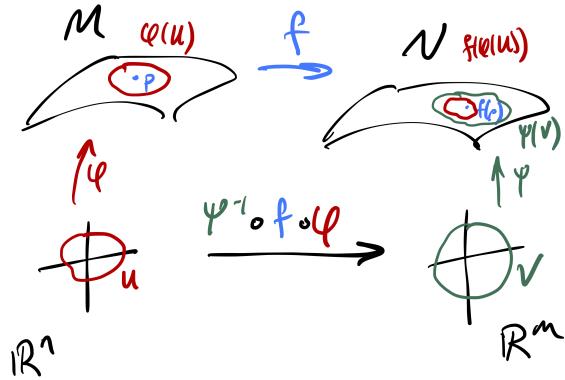


Figure 1.3: Locally converting a map between manifolds to a map between open sets in Euclidean spaces, where we know what differentiability means.

1.2 Tangent Vectors

Notice that there's nothing in [Definition 1.1.1](#) that says that a manifold has to live inside some bigger Euclidean space. This is in contrast to a typical undergraduate differential geometry course (like [MATH 474](#) here at CSU), which is typically focused on surfaces in \mathbb{R}^3 .

Of course, many manifolds (like spheres) do naturally live in some Euclidean space, and it turns out that the Whitney embedding theorem [[27](#), [28](#)] guarantees that all manifolds *can* be embedded in some Euclidean space, but this embedding is not necessarily going to be pleasant to work with. If you've encountered them before, it is often much easier to work with projective spaces and Grassmannians *without* embedding them anywhere in particular.

Since our manifolds don't necessarily live inside a Euclidean space, we have to be a bit careful about what a tangent vector at a point is supposed to be. In particular, a tangent vector at a point lives in a different universe than the point itself: the point is a point in the manifold, but the tangent vector does *not* live on the manifold. This is clear even on a sphere in Euclidean space: the tangent vector to a point on the sphere does not live on the sphere! However, in Euclidean space we can kind of cheat and think of both the point and the tangent vector as both living inside the ambient Euclidean space. In general, we can't get away with this.

Now even in Euclidean space you have to be a little careful with this mixing of point and tangent vector: the tangent vector can't be any *arbitrary* vector in the ambient Euclidean space: it has to lie in the *tangent space* at the point, which we usually visualize as some plane which is tangent to the sphere at the point. But if we're not in Euclidean space, things are even worse: if there's supposed to be some subspace which is "tangent at a point," where does it even live? Surely not in the manifold itself, but we're not thinking of the manifold as sitting inside some bigger space, so there's no "outside" where it can be.

This is a surprisingly nontrivial issue, requiring us to construct some abstract vector space which doesn't really live anywhere in particular. The construction is fairly non-obvious, and feels like a sneaky trick the first few times you encounter it. This is one of those situations where it seems like you're turning a concept inside-out; at least for me, the first time I encountered the following way of thinking, it made me feel uncomfortable in a similar way to when I first encountered the natural embedding of a vector space into its double dual. I will say that at some point my brain switched from "this is weird and awkward" to "this is

obviously the right way to do it" and I think most differential geometers have had similar experiences, so this is something you can eventually develop intuition for.

Given that preamble, what's the idea? We're going to work by analogy with a way of thinking about vectors in \mathbb{R}^n that's slightly different from what you may be used to. In words, we'll identify a vector $\vec{v} \in \mathbb{R}^n$ with the operator on differentiable functions which gives the directional derivative (in the direction of \vec{v}) of a function.

That's a little vague, so let's try to characterize a tangent vector \vec{v} to a point $p \in \mathbb{R}^n$ in this way (here I'm using the notation p rather than \vec{p} , because I'm just thinking of p as a point in a manifold, not as an element of a vector space). Given \vec{v} , I claim we can find some smooth curve $\alpha : (-\epsilon, \epsilon) \rightarrow \mathbb{R}^n$ with $\alpha(0) = p$ and $\alpha'(0) = \vec{v}$; see Figure 1.4. In coordinates, if

$$\alpha(t) = (x_1(t), \dots, x_n(t)) \quad \text{for } t \in (-\epsilon, \epsilon),$$

where the coordinate functions $x_i : (-\epsilon, \epsilon) \rightarrow \mathbb{R}$ are themselves smooth, then

$$\alpha'(0) = (x'_1(0), \dots, x'_n(0)) = \vec{v}.$$



Figure 1.4: A curve through p with velocity \vec{v} at p .

(In this specific case, we could take $\alpha(t) = p + t\vec{v}$, but it turns out not to matter which curve satisfying $\alpha(0) = p$ and $\alpha'(0) = \vec{v}$ we take.)

Now, say $f : U \rightarrow \mathbb{R}$ is differentiable, where $U \subseteq \mathbb{R}^n$ is some neighborhood of p . Then the directional derivative of f at p in the direction of \vec{v} is

$$\frac{d(f \circ \alpha)}{dt} \Big|_{t=0} = \sum_{i=1}^n \frac{\partial f}{\partial x_i} \Big|_p \frac{dx_i}{dt} \Big|_{t=0} = \left(\sum_{i=1}^n x'_i(0) \frac{\partial}{\partial x_i} \right) f$$

by the Chain Rule.

On the right hand side above, we've written the directional derivative as an operator $L = \sum_{i=1}^n x'_i(0) \frac{\partial}{\partial x_i}$ acting on f . Moreover, this operator depends uniquely on \vec{v} and is a *linear derivation*, meaning that:

- (i) $L(f + \lambda g) = L(f) + \lambda L(g)$ for all f and g differentiable in a neighborhood of p and all $\lambda \in \mathbb{R}$;

(ii) $L(fg) = f(p)L(g) + g(p)L(f)$ (i.e., the Product Rule).

If it's not clear to you, it's worth looking at the above and convincing yourself that the operator L didn't really depend on the choice of α : it really only depends on p and \vec{v} (you may need to go back to the justification from MATH 517 that the directional derivative is well-defined).

The upshot is that, given a point p and a tangent vector \vec{v} at p , we get a directional derivative operator L . And, conversely, if we know how to compute a directional derivative, we (at least implicitly) know the direction, so this is really a bijective correspondence.

The benefit of thinking in this way is that we can talk about differential operators like L on any manifold; after all, manifolds locally look like \mathbb{R}^n by definition. So, with that long preamble in mind, here's the definition of a tangent vector:

Definition 1.2.1. Let M^n be a manifold. A smooth function $\alpha: (-\epsilon, \epsilon) \rightarrow M$ is a (smooth) curve in M . Suppose $\alpha(0) = p \in M$ and let \mathcal{D}_p be the set of functions on M that are differentiable in a neighborhood of p . The *tangent vector* to a curve α at $t = 0$ is a function $\alpha'(0): \mathcal{D}_p \rightarrow \mathbb{R}$ given by

$$\alpha'(0)f := \left. \frac{d(f \circ \alpha)}{dt} \right|_{t=0} = (f \circ \alpha)'(0).$$

A *tangent vector at p* is the tangent vector at $t = 0$ of some curve $\alpha: (-\epsilon, \epsilon) \rightarrow M$ with $\alpha(0) = p$.

The set of all tangent vectors at p is the *tangent space* to M at p , denoted $T_p M$.

This is a nice coordinate-free way of defining things, but it's not very useful for computations. We usually *do* want to work in coordinates for computations (and certainly for any computations that we want to do on the computer), so let's see what all this means in local coordinates.

Say that (U, ϕ) is a coordinate chart containing $p \in M$ so that $\phi(\vec{0}) = p$, that $\alpha: (-\epsilon, \epsilon) \rightarrow M$ is smooth with $\alpha(0) = p$, and that f is a differentiable function in a neighborhood of p . Then

$$(\phi^{-1} \circ \alpha)(t) = (x_1(t), \dots, x_n(t))$$

for some smooth functions $x_1, \dots, x_n: (-\epsilon, \epsilon) \rightarrow \mathbb{R}$. Then

$$\alpha'(0)f = \left. \frac{d(f \circ \alpha)}{dt} \right|_{t=0} = \left. \frac{d}{dt}(f \circ \phi(x_1(t), \dots, x_n(t))) \right|_{t=0} = \left. \sum_{i=1}^n x'_i(0) \frac{\partial f}{\partial x_i} \right|_{\vec{0}} = \left(\sum_{i=1}^n x'_i(0) \frac{\partial}{\partial x_i} \Big|_{\vec{0}} \right) f,$$

where I'm using a very common abuse of notation in the second expression to think of f as being a function on the coordinate chart U , and hence a function of coordinates x_1, \dots, x_n (of course, it's really $f \circ \phi$ which is a function of x_1, \dots, x_n , as we see in the middle expression).

This all means that we can write the tangent vector $\alpha'(0) \in T_p M$ in local coordinates as

$$\alpha'(0) = \sum_{i=1}^n x'_i(0) \left(\frac{\partial}{\partial x_i} \Big|_{\vec{0}} \right).$$

Since the $x'_i(0)$ are just scalars, what we're doing here is writing $\alpha'(0)$ in terms of the *local coordinate basis* $\left\{ \left(\frac{\partial}{\partial x_1} \Big|_{\vec{0}}, \dots, \frac{\partial}{\partial x_n} \Big|_{\vec{0}} \right) \right\}$ for $T_p M$ associated to the chart (U, ϕ) .

Remark 1.2.2. In practice, we will almost always drop the subscript $\vec{0}$ and just write the basis as $\left\{\frac{\partial}{\partial x_1}, \dots, \frac{\partial}{\partial x_n}\right\}$ and generic tangent vectors as $\sum_{i=1}^n a_i \frac{\partial}{\partial x_i}$.

Exercise 1.2.3. Suppose a point $p \in M$ lies in two different coordinate charts. How are the two different local coordinate bases related?

It's extremely important to keep in mind that, if p and q are distinct points on M , then the tangent spaces $T_p M$ and $T_q M$ are *completely different vector spaces* that in principle have nothing to do with each other (they're both vector spaces of the same dimension, and hence abstractly isomorphic, but that's essentially all we can know without more detailed information about the geometry of M).

Nonetheless, we often want to talk about all tangent spaces at once, so we smush them together:

Definition 1.2.4. The *tangent bundle* of a manifold M , denoted TM , is the (disjoint) union of tangent spaces

$$TM := \bigsqcup_{p \in M} T_p M.$$

Likewise, if $(T_p M)^*$ is the dual of $T_p M$, then the *cotangent bundle* is the union of cotangent spaces

$$T^* M := \bigsqcup_{p \in M} (T_p M)^*.$$

Notice that there are natural projections $\pi : TM \rightarrow M$ and $\tilde{\pi} : T^* M \rightarrow M$ which just record the base point; in other words, π sends a tangent vector at a point to the point (formally, TM and $T^* M$ are *vector bundles*, and this projection is part of their definition).

Theorem 1.2.5. If M is an n -dimensional manifold, then TM and $T^* M$ are $2n$ -dimensional manifolds.

Exercise 1.2.6. Prove Theorem 1.2.5.

$T^* M$ is, in some sense, the most basic example of a *symplectic manifold*. In physics and dynamical systems language, if M is the *configuration space*³ of a (classical) system, then $T^* M$ is *phase space* or *position-momentum space*: this is the natural setting of Hamiltonian mechanics.

Example 1.2.7. $T\mathbb{R}^n \cong \mathbb{R}^n \times \mathbb{R}^n \cong \mathbb{R}^{2n}$.

Example 1.2.8. $TS^1 \cong S^1 \times \mathbb{R}$, the infinite cylinder.

Example 1.2.9. TS^2 is a *non-trivial* bundle over S^2 , meaning that it is *not* homeomorphic to $S^2 \times \mathbb{R}^2$. In particular, one can show that US^2 , which is the *unit tangent bundle* (the subset of the tangent bundle consisting only of unit tangent vectors) is homeomorphic to $\text{SO}(3) \cong \mathbb{RP}^3$, the real projective space, which is a circle bundle over S^2 different from the trivial circle bundle $S^2 \times S^1$. (For those that have taken graduate topology, $\pi_1(S^2 \times S^1) \cong \mathbb{Z}$, whereas $\pi_1(\mathbb{RP}^3) \cong \mathbb{Z}/2\mathbb{Z}$.)

³Meaning it records the positions of particles; for example if we're doing dynamics of n points on the circle, the configuration space is the n -torus $S^1 \times \dots \times S^1 = (S^1)^n$.

1.3 The Differential

In multivariable calculus, any differentiable map $f: \mathbb{R}^m \rightarrow \mathbb{R}^n$ has a corresponding differential df .⁴ While in a multivariable calculus setting we often think of df as a map $\mathbb{R}^m \rightarrow \mathbb{R}^n$ as well (meaning it would have the same domain and codomain as f), or even more concretely as an $n \times m$ matrix, this is misleading.

In fact, if you were reading the first sentence of the previous paragraph very carefully, you may have noticed that it was not quite correct. Really, I should have said that any differentiable map has a corresponding differential *at a point p* . There is no single “differential” for a non-linear map: we get different differentials in the neighborhood of each point in the domain.

So more accurately, if p is in the domain of f , then we have a differential df_p at p . Now, what is the domain of df_p ? Recall, df_p is supposed to be the best linear approximation of f at p , so we’re supposed to think of the inputs to df_p as being slight perturbations of p . In other words, the domain of df_p is really the tangent space $T_p \mathbb{R}^m$ (which is in a very precise sense the space of *infinitesimal* perturbations of p).

Now, $T_p \mathbb{R}^m$ is isomorphic to \mathbb{R}^m , but not just in some abstract sense: the tangent bundle $T \mathbb{R}^m$ is trivial, meaning that $T \mathbb{R}^m \cong \mathbb{R}^m \times \mathbb{R}^m$, and a concrete trivialization is given by parallel translating a tangent vector from being based at p to being based at the origin. This is the sense in which we can think of the domain of each df_p as being the “same” \mathbb{R}^m .

But if M is some more general m -dimensional manifold, then, while any tangent space $T_p M$ is still abstractly isomorphic to \mathbb{R}^m , the tangent bundle is likely not to be trivial, so we can’t make any general identifications of different tangent spaces with the same \mathbb{R}^m . So if we want to generalize the notion of differential to manifolds, we need to be careful, even when thinking about functions on \mathbb{R}^m , to maintain the distinction between \mathbb{R}^m (thought of an m -dimensional manifold) and $T_p \mathbb{R}^m$.

And what about the codomain of df_p ? Again, we should think more conceptually about what the differential really does. Like all derivatives, the differential tells us something about how the change in an input to a function changes the output of the function. So the inputs to the differential will be tangent vectors at some point in the domain of f (recording an infinitesimal change to the input, or equivalently an initial position [the point] and velocity [the tangent vector] of some path), and the outputs will be tangent vectors at the corresponding point in the range (recording the infinitesimal change in the output). Notice that $T_q \mathbb{R}^n$ is again isomorphic with \mathbb{R}^n by parallel translation.

In other words, given a differentiable map $f: \mathbb{R}^m \rightarrow \mathbb{R}^n$ and a point p in the domain of f , the differential of f at p is a map $df_p: T_p \mathbb{R}^m \rightarrow T_{f(p)} \mathbb{R}^n$, where domain and range are usually identified with \mathbb{R}^m and \mathbb{R}^n in a standard way. This, then, is the way of thinking about differentials which generalizes.

Definition 1.3.1. Suppose M^m and N^n are smooth manifolds and $f: M \rightarrow N$ is smooth. For each $p \in M$, define the *differential* of f at p , denoted $df_p: T_p M \rightarrow T_{f(p)} N$, as follows: For any $v \in T_p M$, choose a smooth curve $\alpha: (-\epsilon, \epsilon) \rightarrow M$ such that $\alpha(0) = p$ and $\alpha'(0) = v$. Let $\beta = f \circ \alpha$, which is a smooth curve in N . Then

$$df_p(v) := \beta'(0).$$

Lemma 1.3.2. *This is a well-defined linear map.*

Exercise 1.3.3. Prove Lemma 1.3.2.

In the special case that $f: M \rightarrow \mathbb{R}$ is a smooth function, the differential of f applied to a tangent vector v is the same thing as computing the derivative of f in the direction of v :

⁴You may have encountered this under the name *Jacobian* rather than *differential*; in the case $n = 1$, this is [more or less] the gradient of the function.

Lemma 1.3.4. Suppose M is a manifold, $p \in M$, $v \in T_p M$, and that $f: M \rightarrow \mathbb{R}$ is differentiable in a neighborhood of p . Then

$$vf = df_p v.$$

Proof. This is just a matter of unwinding definitions. Let $\alpha: (-\epsilon, \epsilon) \rightarrow M$ be a smooth curve so that $\alpha(0) = p$ and $\alpha'(0) = v$. Then by definition the left hand side is

$$(vf)(p) = (\alpha'(0)f)(p) = (f \circ \alpha)'(0).$$

But this is exactly the definition of $df_p v$. \square

Let's see what this means in local coordinates, which is how we'll actually compute in practice. Suppose $p \in M$ and (V, ψ) is a local coordinate chart on N containing $f(p)$. By Definition 1.1.6, there exists some compatible chart (U, ϕ) on M containing p so that $f(\phi(U)) \subseteq \psi(V)$. Then the curve α on M (or, really, its restriction to $\phi(U)$) gives a corresponding curve

$$(x_1(t), \dots, x_m(t)) = \phi^{-1} \circ \alpha(t)$$

in $U \subseteq \mathbb{R}^m$, and similarly the curve β on N has a corresponding curve

$$(y_1(t), \dots, y_n(t)) = \psi^{-1} \circ \beta(t).$$

See Figure 1.5.

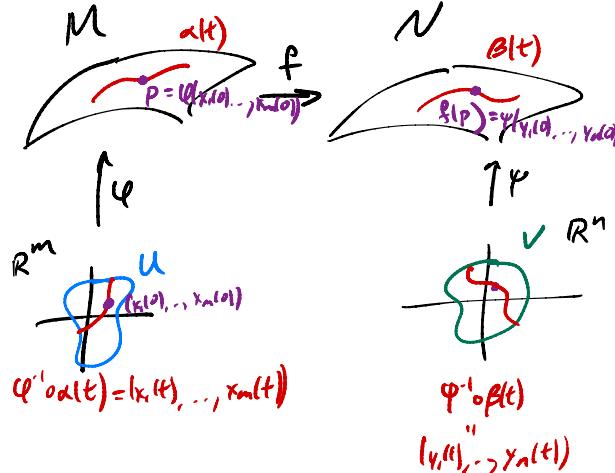


Figure 1.5: The differential in local coordinates

Of course,

$$\psi^{-1} \circ \beta(t) = \psi^{-1} \circ f \circ \alpha(t) = \psi^{-1} \circ f \circ \phi(x_1(t), \dots, x_m(t)),$$

so we can think of the y_i as functions of the x_j , and the curve in V can be written as

$$(y_1(x_1(t), \dots, x_m(t)), \dots, y_n(x_1(t), \dots, x_m(t))).$$

Working in local coordinates means doing the computation at the level of the map $\psi^{-1} \circ f \circ \phi: U \subseteq \mathbb{R}^m \rightarrow V \subseteq \mathbb{R}^n$, which has a corresponding differential

$$d(\psi^{-1} \circ f \circ \phi)_{(x_1(0), \dots, x_m(0))}: T_{(x_1(0), \dots, x_m(0))}\mathbb{R}^m \rightarrow T_{(y_1(0), \dots, y_m(0))}\mathbb{R}^n.$$

By definition, evaluating this differential on the tangent vector $(x'_1(0), \dots, x'_m(0)) \in T_{(x_1(0), \dots, x_m(0))}\mathbb{R}^m$ gives

$$\begin{aligned} \frac{d}{dt} \Big|_{t=0} (y_1(x_1(t), \dots, x_m(t)), \dots, y_n(x_1(t), \dots, x_m(t))) &= \left(\sum_{j=1}^m \frac{\partial y_1}{\partial x_j} \Big|_x \frac{dx_j}{dt} \Big|_{t=0}, \dots, \sum_{j=1}^m \frac{\partial y_n}{\partial x_j} \Big|_x \frac{dx_j}{dt} \Big|_{t=0} \right) \\ &= \left(\sum_{j=1}^m \frac{\partial y_1}{\partial x_j} \Big|_x x'_j(0), \dots, \sum_{j=1}^m \frac{\partial y_n}{\partial x_j} \Big|_x x'_j(0) \right) \\ &= \begin{bmatrix} \frac{\partial y_1}{\partial x_1} & \cdots & \frac{\partial y_1}{\partial x_m} \\ \vdots & \ddots & \vdots \\ \frac{\partial y_n}{\partial x_1} & \cdots & \frac{\partial y_n}{\partial x_m} \end{bmatrix} \begin{bmatrix} x'_1(0) \\ \vdots \\ x'_m(0) \end{bmatrix} \\ &= \begin{bmatrix} \frac{\partial y_i}{\partial x_j} \end{bmatrix}_{i,j} x'(0), \end{aligned}$$

where $\begin{bmatrix} \frac{\partial y_i}{\partial x_j} \end{bmatrix}_{i,j}$ is the $n \times m$ matrix of partials (i.e., the Jacobian).

So when we work in local coordinates, this is just standard multivariable calculus, as you would hope.

Just to reiterate, the differential (at a point) of a map between manifolds is going to input a tangent vector at a point in the domain and output a tangent vector at a point in the range. As a special case, suppose we have a smooth function on a manifold M ; that is, a smooth map $f: M \rightarrow \mathbb{R}$. Then, at a point $p \in M$, the differential $df_p: T_p M \rightarrow T_{f(p)}\mathbb{R}$. Now, the tangent bundle to \mathbb{R} is trivial, so we can identify $T_{f(p)}\mathbb{R}$ with \mathbb{R} in a canonical way, and therefore think about df_p as a linear map $T_p M \rightarrow \mathbb{R}$.

In other words, by way of this identification of $T_{f(p)}\mathbb{R}$ with \mathbb{R} , we can think of df_p as a linear functional on the vector space $T_p M$. Or, in fancier language, df_p is an element of the dual space $(T_p M)^*$. We'll come back to this later when we start talking about differential forms, but this gives a hint of why we might care about cotangent spaces and the cotangent bundle.

Just as in multivariable calculus, when we have a smooth map $f: M \rightarrow N$, we can talk about critical points and critical values, and doing so is often quite important.

Definition 1.3.5. Let $f: M \rightarrow N$ be smooth. A point $p \in M$ is a *critical point* of f if $df_p: T_p M \rightarrow T_{f(p)}N$ is not surjective; then $f(p)$ is a *critical value* of f . A point $q \in N$ which is not a critical value is a *regular value* of f .

Example 1.3.6 (Cliché Example). Consider the function f on the torus depicted in Figure 1.6. In words, f is the height function on this upright torus that just records the z -coordinate. I've shown the critical values in red, as well as a couple of regular values in green. Also, back on the torus I've shown the inverse images of the critical and regular values, as well as some tangent vectors to regular points and their images under the differential (in blue and purple), to hopefully convince you the differential really is surjective at these points.

Notice that there are basically three different types of critical points: a minimum, a maximum, and two saddle points. The differential can't possibly be surjective at a (local) minimum: there's no direction you



Figure 1.6: The height function on a torus.

could go which will cause the function to decrease, so the image of the differential has to miss an entire half of its codomain (and therefore, by linearity, everything except the origin). It's somewhat less geometrically clear that the saddle points are critical points, though in this case it's fairly easy to see once you realize the tangent plane to one of the saddle points is horizontal.

A couple of other things to notice about this picture:

- (i) The inverse images of the two critical points coming from the saddle points contain plenty of regular points: indeed, the differential is surjective at any of the points in the inverse image *except* the saddle point.
- (ii) The level sets of regular values are smooth (collections of) curves, either a single circle or two disjoint circles; in other words, they are smooth (possibly disconnected) 1-dimensional manifolds. The level sets of the critical points are either a single point (the minimum and maximum) or a wedge of two circles (like an ∞ symbol). In both cases, the result is not a smooth 1-manifold: a point is a 0-manifold and the wedge of two circles is not locally like a copy of \mathbb{R} near the point where the two circles meet (which of course is exactly the saddle point).

Remark 1.3.7. I call this a cliché example because it is the first picture everybody draws when they talk about Morse theory, which says that the topology of a manifold is in some sense encoded in the critical points of any sufficiently generic function on the manifold. Such functions are called *Morse functions*, and the function in this example is a Morse function. In particular, Morse theory tells you that if you are traversing the manifold “up” according to a Morse function, the topology only changes when you pass a critical point. In this example, the topology changes from the empty set to a disk when we pass the minimum, then from

a disk to a

(which is homotopic to a circle) when we pass the first saddle point, then to a punctured

torus (which is homotopic to a wedge of circles) when we pass the second saddle point, and then finally the puncture is filled in when we pass the maximum. Something analogous happens in general.

Example 1.3.8. Consider the function $f(x, y, z) = z^2$ on the sphere (see Figure 1.7). Again, I've shown critical values in red and orange and a regular value in green, and the corresponding level sets on the domain, as well as some tangent vectors to regular points and their images under the differential.



Figure 1.7: A smooth function on the sphere which does not have isolated critical points.

Some new features:

- (i) The critical points aren't all isolated: the entire equator consists of critical points (which are all global minima).
- (ii) The inverse images of critical values aren't necessarily connected: the north and south poles both map to 1.

Notice again that the level sets of regular values are (collections of) smooth curves. In this case one critical level set is also a smooth curve, and the other is not a curve at all.

(This is not a Morse function because not all critical points are isolated, but it is a *Morse–Bott function*, which is almost as good.)

After looking at these examples, hopefully the following theorem suggests itself:

Theorem 1.3.9 (Level Set Theorem). *If $m \geq n$, $f: M^m \rightarrow N^n$ is smooth, and $q \in N$ is a regular value of f , then $f^{-1}(q)$ is a smooth submanifold of M of dimension $m - n$.*

This theorem is basically an application of the Inverse Function Theorem, but before we work on proving it, let's look at some more examples.

Example 1.3.10. Define $f: \mathbb{R}^n \rightarrow \mathbb{R}$ by $f(x_1, \dots, x_n) = x_1^2 + \dots + x_n^2$. Then

$$df_{(x_1, \dots, x_n)} = \begin{bmatrix} \frac{\partial f}{\partial x_1} & \cdots & \frac{\partial f}{\partial x_n} \end{bmatrix} = [2x_1 \quad \cdots \quad 2x_n]$$

only fails to be full rank at the origin, so 0 is the only critical value, and for all $r > 0$ the level set $f^{-1}(r)$ is a smooth manifold of dimension $n - 1$. But $f^{-1}(r)$ is nothing but the sphere of radius \sqrt{r} , so we've just given an alternative (and somehow conceptually prior) proof that spheres are manifolds.

Example 1.3.11 (Relevant to frame theory). Let $\text{Mat}_{d \times N}(\mathbb{C})$ be the space of $d \times N$ complex matrices. This is trivially a $2dN$ -dimensional manifold, since $\text{Mat}_{d \times N}(\mathbb{C}) \cong \mathbb{C}^{dN}$, which is just a copy of \mathbb{R}^{2dN} if you ignore the complex structure. Now, let $\mathcal{H}(d)$ be the space of $d \times d$ Hermitian matrices (i.e., $d \times d$ complex matrices A so that $A^* = A$, where A^* is the conjugate transpose of A) and define the map $\Phi: \text{Mat}_{d \times N}(\mathbb{C}) \rightarrow \mathcal{H}(d)$ by

$$\Phi(A) = AA^*.$$

(In frame theory, we would think of $A \in \text{Mat}_{d \times N}(\mathbb{C})$ as a frame and AA^* as the corresponding frame operator.)

I claim that the identity matrix $I_{d \times d} \in \mathcal{H}(d)$ is a regular value of Φ (**Exercise:** Prove this); assuming this, the theorem tells us that $\Phi^{-1}(I_{d \times d})$ is a smooth manifold of dimension

$$2dN - d^2 = d(2N - d).$$

(Notice that $A \in \Phi^{-1}(I_{d \times d})$ means that the rows of A are Hermitian orthonormal. Since each row is a vector in \mathbb{C}^N , this means that we can think of $\Phi^{-1}(I_{d \times d})$ as the space of all [ordered] d -tuples of Hermitian orthonormal vectors in \mathbb{C}^N . This space is an example of a *Stiefel manifold*, and usually denoted $\text{St}_d(\mathbb{C}^N)$ or $V_d(\mathbb{C}^N)$. In frame theory, $\Phi^{-1}(I_{d \times d})$ is precisely the space of *Parseval frames*.)

Example 1.3.12. Let $N = d$ in the previous example. Then $\Phi^{-1}(I_{d \times d})$ is a smooth manifold of dimension d^2 that consists of those $d \times d$ complex matrices A so that $AA^* = I_{d \times d}$. But this is precisely the unitary group $U(d)$! So we've proved that $U(d)$ is a d^2 -dimensional manifold for any d .

Remark 1.3.13. You can play the same game over \mathbb{R} to show that real Stiefel manifolds and the orthogonal group $O(d)$ are manifolds.

1.4 Immersions and Embeddings

Let's work up to proving [Theorem 1.3.9](#), including defining some more terminology.

Definition 1.4.1. Suppose $f: M \rightarrow N$ is smooth. Then f is an *immersion* if df_p is injective for all $p \in M$ (note that this implies $\dim(M) \leq \dim(N)$).

If f is also a homeomorphism onto its image (continuous bijection with continuous inverse), then f is an *embedding*. The image of an embedding is a *submanifold*.

If f is bijective and f^{-1} is smooth, then f is a *diffeomorphism*. f is called a *local diffeomorphism* at $p \in M$ if there exist neighborhoods U of p and V of $f(p)$ so that $f: U \rightarrow V$ is a diffeomorphism.

Finally, we say that f is a *submersion* if df_p is surjective for all $p \in M$.

Example 1.4.2. Define $\alpha: \mathbb{R} \rightarrow \mathbb{R}^2$ by $\alpha(t) = (t^3, t^2)$ (see [Figure 1.8](#)).

This is not an immersion because $d\alpha_0 = \begin{bmatrix} \alpha'_1(0) \\ \alpha'_2(0) \end{bmatrix} = \begin{bmatrix} 0 \\ 0 \end{bmatrix}$ has rank 0. Intuitively, the problem is that the velocity is zero when $t = 0$, even though the map overall is continuous and injective.

Example 1.4.3. Define $\beta: \mathbb{R} \rightarrow \mathbb{R}^2$ by $\beta(t) = (t^3 - 4t, t^2 - 4)$, which is a deformation of the previous example (see [Figure 1.9](#)).

Now the differential is given by

$$d\beta_t = \begin{bmatrix} 3t^2 - 4 \\ 2t \end{bmatrix},$$



Figure 1.8: The image of the map α from Example 1.4.2. The fact that it has a cusp suggests that it will fail to be an immersion, despite being injective and smooth.

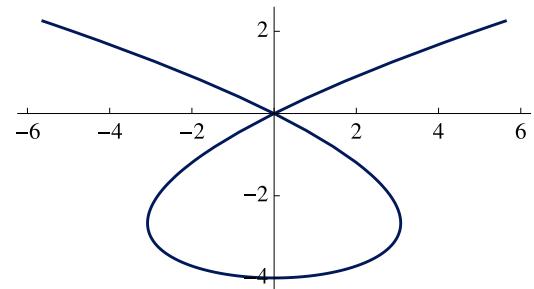


Figure 1.9: The image of the map β from Example 1.4.3. This map is an immersion, but not an embedding.

which is never the zero matrix since the second entry being 0 implies $t = 0$, and hence that the first entry is -4 . Hence, $d\beta_t$ is always rank 1, and hence always injective, so β is an immersion.

However, β is not an embedding because it is not injective: $\beta(-2) = (0, 0) = \beta(2)$, so there is a double point at the origin.

Example 1.4.4. Suppose (U, ϕ) is a local coordinate chart on a manifold M containing a point $p \in M$. Then $\phi: U \subset \mathbb{R}^n \rightarrow M$ is a local diffeomorphism at $\phi^{-1}(p) \in U$. This just follows from the definition of a coordinate chart (Definition 1.1.1): ϕ maps U bijectively onto $\phi(U)$, which is an open neighborhood of p , and both ϕ and ϕ^{-1} are smooth. (To be really pedantic, (U, id) gives a global coordinate chart for the n -manifold U , and then, following Definition 1.1.6, $\phi: U \rightarrow \phi(U)$ is smooth because $\phi^{-1} \circ \phi \circ \text{id} = \text{id}$ is smooth everywhere on U . Similarly, $\phi^{-1}: \phi(U) \rightarrow U$ is smooth because $\text{id} \circ \phi^{-1} \circ \phi = \text{id}$ is smooth everywhere on U .)

If $f: M \rightarrow N$ is a local diffeomorphism at $p \in M$, then $df_p: T_p M \rightarrow T_{f(p)} N$ is a linear isomorphism (i.e., invertible linear map). Perhaps somewhat surprisingly, the converse is also true. This is the appropriate generalization of the Inverse Function Theorem to the manifold setting, and the proof essentially involves applying the Inverse Function Theorem in local coordinates:

Proposition 1.4.5. Suppose $f: M \rightarrow N$ is smooth. If $p \in M$ and $df_p: T_p M \rightarrow T_{f(p)} N$ is an isomorphism, then f is a local diffeomorphism at p .

Proof. We are going to apply the Inverse Function Theorem (Theorem 1.4.7 below) to $\psi^{-1} \circ f \circ \phi: \phi(U) \rightarrow \psi(V)$, where (U, ϕ) and (V, ψ) are the coordinate charts guaranteed to exist by the definition of smoothness (Definition 1.1.6).

Notice that, by the Chain Rule,

$$d(\psi^{-1} \circ f \circ \phi)_{\phi^{-1}(p)} = d\psi_{f(p)}^{-1} \circ df_p \circ d\phi_{\phi^{-1}(p)}.$$

But then we already know that $d\phi$ and $d\psi^{-1}$ are isomorphisms (since coordinate charts are local diffeomorphisms [Example 1.4.4]), so df_p being an isomorphism implies that $d(\psi^{-1} \circ f \circ \phi)_{\phi^{-1}(p)}$ is also an isomorphism.

But then the Inverse Function Theorem implies $\psi^{-1} \circ f \circ \phi$ is a local diffeomorphism at $\phi^{-1}(p)$. Since ψ and ϕ^{-1} are local diffeomorphisms (again, Example 1.4.4) and compositions of local diffeomorphisms are local diffeomorphisms, it follows that

$$\psi \circ (\psi^{-1} \circ f \circ \phi) \circ \phi^{-1} = (\psi \circ \psi^{-1}) \circ f \circ (\phi \circ \phi^{-1}) = f$$

is also a local diffeomorphism. □

Here's a statement of the Inverse Function Theorem in the language of differentials and local diffeomorphisms. It is equivalent to the usual statement you would see in a multivariable calculus or analysis course.

Exercise 1.4.6. Convince yourself of the previous sentence.

Theorem 1.4.7 (Inverse Function Theorem). *If $F: U \subseteq \mathbb{R}^n \rightarrow \mathbb{R}^n$ is given by $(x_1, \dots, x_n) \mapsto (y_1, \dots, y_n)$, then $dF_p = \left[\frac{\partial y_i}{\partial x_j} \Big|_p \right]_{i,j}$ being nonsingular at $p \in U$ implies that F is a local diffeomorphism at p .*

We're now ready to prove the Level Set Theorem (Theorem 1.3.9):

Proof of Theorem 1.3.9. Recall that, in the statement, we're assuming that $f: M \rightarrow N$ is smooth and that $q \in N$ is a regular value of f , and the goal is to show that $f^{-1}(q)$ is a smooth submanifold of M of dimension $m - n$.

Suppose (U, ϕ) is a coordinate chart centered at $p \in f^{-1}(q)$ ⁵ and (V, ψ) is a chart at q . See Figure 1.10. Then define the map

$$g := \psi^{-1} \circ f \circ \phi : U \rightarrow \mathbb{R}^n.$$

By assumption,

$$dg_{\vec{0}} = d(\psi^{-1} \circ f \circ \phi)_{\vec{0}} = (d\psi^{-1})_{f(p)} \circ df_p \circ d\phi_{\vec{0}}$$

is surjective since df_p is and the other two terms are isomorphisms.



Figure 1.10: A sketch of the key neighborhoods and maps in the proof of Theorem 1.3.9.

Hence, after a linear change of coordinates $dg_{\vec{0}}$ can be written as the $n \times m$ block matrix $[I_{n \times n} \quad 0_{n \times (m-n)}]$. Define

$$G(x_1, \dots, x_m) = (g(x_1, \dots, x_n), x_{n+1}, \dots, x_m),$$

which has differential $dG_{\vec{0}} = I_{m \times m}$ in these coordinates. By the Inverse Function Theorem, G is a local diffeomorphism at $\vec{0}$ and by construction $g \circ G^{-1}$ is the standard projection onto the first n coordinates.

So in these coordinates $f^{-1}(q) \cap \phi(U) = \phi(0, \dots, 0, x_{n+1}, \dots, x_m)$, so x_{n+1}, \dots, x_m give local coordinates near p for $f^{-1}(q)$. We can do the same at any other point of $f^{-1}(q)$, so this gives a system of coordinate charts on $f^{-1}(q)$, and the transition maps on overlapping charts are smooth because they are just restrictions (or projections, if you prefer) of the corresponding transition maps for the charts on M . \square

⁵Meaning that $p = \phi(\vec{0})$.

1.5 Vector Fields

Now that we have defined tangent vectors and seen how to push them around with differentials, the next natural object to try to define is a vector field. In physical problems, a vector field might be the velocity field of a fluid flow, or an electrical or magnetic field. In both applied and pure problems, we are very often interested in vector fields that arise as the gradients of functions, whether they be Morse functions on an abstract manifold or the energy of conformations in some conformation space. In symplectic geometry we are very interested in Hamiltonian vector fields associated to functions, which are a sort of symplectic gradient. Whereas the gradient is perpendicular to level sets of the function, Hamiltonian vector fields are parallel to level sets, so the function is constant on orbits, which is desirable when the function is an energy function.

So what is a vector field? Intuitively, it's just what you would expect: a choice of tangent vector at each point in the manifold. As with everything in this class, we're mostly interested in *smooth* vector fields, meaning that the tangent vector should in some sense vary smoothly as you move around the manifold.

The preceding sentence hopefully makes sense on an intuitive level, but you should stop and think about how you might try to make it into a rigorous definition. Unless you've already seen the forthcoming definition before, it's probably not so easy.

The problem is that, as alluded to previously, different tangent spaces are not directly comparable. How do you compare $v_1 \in T_{p_1} M$ with $v_2 \in T_{p_2} M$ and, ideally, put them into some difference quotient?

Since we know how to compare tangent spaces in Euclidean space (by parallel translating), one strategy is to work in local coordinates and require the tangent vectors to form a smooth vector field in each local coordinate chart. But this rather unwieldy, and in any case we want to state definitions without reference to local coordinates if at all possible. Hence the following definition, which encapsulates exactly the idea above in a very concise (though admittedly kind of abstract and hard to visualize) way:

Definition 1.5.1. A *vector field* X on a smooth manifold M is a smooth section of the tangent bundle TM .

This requires some unpacking, especially since there's at least one term in this definition which has not been defined yet.

First, recall ([Theorem 1.2.5](#)) that TM is a smooth manifold, so it makes sense to talk about smoothness of a map $X: M \rightarrow TM$. So the first part of [Definition 1.5.1](#) is that a vector field is such a smooth map.

This makes sense: such a map takes as input a point in the manifold and outputs a tangent vector. But it would be nonsensical to output a tangent vector at q if the input is p : the word “section” is how we rule this out.

More precisely, recall that we have a natural projection map $\pi: TM \rightarrow M$ which maps $v \in T_p M$ to the base point p . In general, if $E \xrightarrow{\pi} B$ is a vector bundle, then a *section* of the bundle is a map $\sigma: B \rightarrow E$ so that $\pi \circ \sigma = \text{id}_B$, the identity map on B (in algebraic terms, σ is a right inverse of the projection π).

So when we say that X is a smooth section of the tangent bundle, we mean that $X: M \rightarrow TM$ is a smooth map satisfying $\pi \circ X = \text{id}_M$. In other words, that X assigns each point in M a tangent vector at that point in a smoothly-varying way. Which is exactly what a (smooth) vector field should be!

Notation. We use the notation $\mathfrak{X}(M)$ to denote the $C^\infty(M)$ -module of vector spaces on M .

Remark 1.5.2. (i) If $\phi: U \subseteq \mathbb{R}^n \rightarrow M$ is a local coordinate chart at $p \in M$, then

$$X(p) = \sum_{i=1}^n a_i(p) \frac{\partial}{\partial x_i},$$

where each $a_i: U \rightarrow \mathbb{R}$ is smooth and $\left\{ \frac{\partial}{\partial x_i} \right\}_i$ is the local coordinate basis of $T_p M$ associated to (U, ϕ) . Notice that this is exactly the local coordinate definition of a smooth vector field informally formulated above.

(ii) Since each individual tangent vector is really a directional derivative at a point, a vector field can be interpreted as a differential operator on M : it will input a smooth function and output some new smooth function which records the directional derivative at each point in the direction specified by the tangent vector at that point. In other words, we can also interpret a vector field X on a manifold M as a map $C^\infty(M) \rightarrow C^\infty(M)$ given by $f \mapsto Xf$. In local coordinates, the function Xf is given by

$$(Xf)(p) = \sum_{i=1}^n a_i(p) \left. \frac{\partial f}{\partial x_i} \right|_p,$$

which is indeed a smooth function.

Example 1.5.3. Consider the unitary group $U(d)$ and let $I = I_{d \times d}$ be the identity matrix. I claim that a choice of tangent vector at the identity actually determines a vector field on all of $U(d)$.

To see this, recall that an element of $T_I U(d)$ is a tangent vector, which is to say, the velocity of a smooth curve $\alpha: (-\epsilon, \epsilon) \rightarrow U(d)$ with $\alpha(0) = I$. If we think of $U(d)$ as living inside $\text{Mat}_{d \times d}(\mathbb{C}) \cong \mathbb{C}^{d^2} \cong \mathbb{R}^{2d^2}$, then we can think of any tangent vector $\alpha'(0)$ as a $d \times d$ complex matrix. What restrictions does being tangent to $U(d)$ place on this matrix?

Notice that, for all t , $\alpha(t) \in U(d)$, which by definition means that $\alpha(t)\alpha(t)^* = I$. Since this is the defining equation of $U(d)$, differentiating at $t = 0$ should give precisely the condition for being in $T_I U(d)$:

$$0 = \left. \frac{d}{dt} \right|_{t=0} I = \left. \frac{d}{dt} \right|_{t=0} (\alpha(t)\alpha(t)^*) = \alpha'(0)\alpha(0)^* + \alpha(0)\alpha'(0)^* = \alpha'(0) + \alpha'(0)^*$$

since $\alpha(0) = I = \alpha(0)^*$ (note that we're also using $(\alpha(t)^*)' = \alpha'(t)^*$, which is obvious if you choose a basis and write $\alpha(t)$ as a matrix, but kind of annoying to prove in a coordinate-free way).

So we see that $\alpha'(0)^* = -\alpha'(0)$. In other words, elements of $T_I U(d)$ are precisely the skew-Hermitian matrices.

Next, how do we get from a tangent vector at I to a vector field on all of $U(d)$? Well, we know how to push vector fields around by differentials of maps, so if we had a map $U(d) \rightarrow U(d)$ sending I to a specified element $U \in U(d)$, then its differential would send $\Delta \in T_I U(d)$ to something in $T_U U(d)$. Of course, there's an obvious map sending I to U : the map which left-multiplies elements of $U(d)$ by U .

In symbols, for $U \in U(d)$, define the map $L_U: U(d) \rightarrow U(d)$ to be left-multiplication by U , namely $L_U(A) := UA$. Then certainly

$$(dL_U)_I: T_I U(d) \rightarrow T_U U(d)$$

and so, for any $\Delta \in T_I U(d)$, we get a vector field $X_\Delta \in \mathfrak{X}(U(d))$ on $U(d)$ defined by

$$X_\Delta(U) := (dL_U)_I \Delta.$$

We can make this more explicit by finding a formula for $(dL_U)_I$:

Lemma 1.5.4. If $U \in \mathrm{U}(d)$ and $\Delta \in T_I \mathrm{U}(d)$, then

$$(dL_U)_I \Delta = U\Delta.$$

Exercise 1.5.5. Prove Lemma 1.5.4. The key observation here is that matrix multiplication is linear.

Since $(dL_U)_I : T_I \mathrm{U}(d) \rightarrow T_U \mathrm{U}(d)$ is full rank, it's surjective, so $T_U \mathrm{U}(d)$ consists of matrices of the form $U\Delta$ where Δ is skew-Hermitian.

And now it's clear that, for any $\Delta \in T_I \mathrm{U}(d)$, we get the vector field X_Δ defined by

$$X_\Delta(U) = U\Delta \in T_U \mathrm{U}(d).$$

This is what's called a *left-invariant* vector field on $\mathrm{U}(d)$, since it is invariant under the action of $\mathrm{U}(d)$ on itself by left-multiplication:

$$(dL_U)_V X_\Delta(V) = X_\Delta(UV).$$

Exercise 1.5.6. Check this.

In fact, one can show that all left-invariant vector fields are of this form, so there is an identification between the collection of left-invariant vector fields on $\mathrm{U}(d)$ and $T_I \mathrm{U}(d)$.

- Remark 1.5.7.**
- (i) There was nothing special about $\mathrm{U}(d)$ in Example 1.5.3: we could have used any group G which was a manifold (such groups are called *Lie groups*; more on them later!) and gotten the same correspondence between the tangent space at the identity and left-invariant vector fields. This is important because these are the two standard ways differential geometers think about *Lie algebras*, and this construction shows that they are equivalent.
 - (ii) More generally, the fact that $T_U \mathrm{U}(d)$ consists of matrices of the form $U\Delta$ where Δ is skew-Hermitian means that *every* vector field $X \in \mathfrak{X}(\mathrm{U}(d))$ can be written as

$$X(U) = U\Delta_U,$$

where the skew-Hermitian matrix Δ_U depends on U . So then a vector field on $\mathrm{U}(d)$ induces a mapping $\mathrm{U}(d) \rightarrow T_I \mathrm{U}(d)$ given by $U \mapsto \Delta_U$.

1.5.1 The matrix exponential

This is not directly related to vector fields, but another reason to be interested in $T_I \mathrm{U}(d)$ (or, more generally, the tangent space at the identity of any Lie group) is that it provides local coordinates on almost all of $\mathrm{U}(d)$ by way of the matrix exponential.

First of all, if $\Delta \in T_I \mathrm{U}(d)$ (i.e., Δ is skew-Hermitian), then I claim that $\exp(\Delta) \in \mathrm{U}(d)$, where \exp is the matrix exponential defined by the power series

$$\exp(\Delta) = I + \Delta + \frac{1}{2!}\Delta^2 + \frac{1}{3!}\Delta^3 + \dots$$

In other words, the claim is that $\exp : T_I \mathrm{U}(d) \rightarrow \mathrm{U}(d)$.

To see that $\exp(\Delta) \in \mathrm{U}(d)$, we need to show that $\exp(\Delta) \exp(\Delta)^* = I$, or equivalently $\exp(\Delta)^* = \exp(\Delta)^{-1}$. Now

$$\begin{aligned}\exp(\Delta)^* &= \left(I + \Delta + \frac{1}{2!} \Delta^2 + \frac{1}{3!} \Delta^3 + \dots \right)^* \\ &= I + \Delta^* + \frac{1}{2!} (\Delta^2)^* + \frac{1}{3!} (\Delta^3)^* + \dots \\ &= I + \Delta^* + \frac{1}{2!} (\Delta^*)^2 + \frac{1}{3!} (\Delta^*)^3 + \dots \\ &= I + (-\Delta) + \frac{1}{2!} (-\Delta)^2 + \frac{1}{3!} (-\Delta)^3 + \dots \\ &= \exp(-\Delta),\end{aligned}$$

and by expanding the product of power series one can show that

$$\exp(\Delta) \exp(\Delta)^* = \exp(\Delta) \exp(-\Delta) = \exp(\Delta - \Delta) = I.$$

(It is absolutely essential in the argument for $\exp(\Delta) \exp(-\Delta) = \exp(\Delta - \Delta)$ that Δ and $-\Delta$ commute. When A and B are matrices with $AB \neq BA$, it is not necessarily true that $\exp(A) \exp(B) = \exp(A + B)$; see the Baker–Campbell–Hausdorff formula in general [16, Chapter 5].)

Claim. $\exp : T_I \mathrm{U}(d) \rightarrow \mathrm{U}(d)$ is surjective.

Proof. If $U \in \mathrm{U}(d)$, then the eigenvalues of U are all unit complex numbers $e^{i\theta_1}, \dots, e^{i\theta_d}$, so U has the spectral decomposition

$$U = V \Theta V^*,$$

where

$$\begin{aligned}\Theta &= \mathrm{diag}(e^{i\theta_1}, \dots, e^{i\theta_d}) \\ &= \mathrm{diag}\left(1 + i\theta_1 + \frac{1}{2!} (i\theta_1)^2 + \dots, \dots, 1 + i\theta_d + \frac{1}{2!} (i\theta_d)^2 + \dots\right) \\ &= \exp(\mathrm{diag}(i\theta_1, \dots, i\theta_d)).\end{aligned}$$

(Note that the set of all such Θ forms a torus; this turns out to be a *maximal torus* inside $\mathrm{U}(d)$).

But then $H = V \mathrm{diag}(i\theta_1, \dots, i\theta_d) V^*$ is skew-Hermitian, and hence in $T_I \mathrm{U}(d)$, and I claim that

$$U = \exp(H).$$

This follows from the more general fact about exponentiating matrix conjugates stated below in [Lemma 1.5.8](#).

Just to verify, let's check this on a random example. Here's a random element of $\mathrm{U}(3)$ (generated in *Mathematica* with `RandomVariate[CircularUnitaryMatrixDistribution[3]]`):

$$U = \begin{bmatrix} -0.392089 + 0.77069i & -0.175305 - 0.0691595i & 0.460085 - 0.0714797i \\ -0.359804 + 0.151554i & 0.618136 + 0.524479i & -0.295949 - 0.32065i \\ 0.103894 + 0.298465i & -0.226492 + 0.505981i & -0.312325 + 0.703749i \end{bmatrix}.$$

Computing the spectral decomposition yields

$$V = \begin{bmatrix} -0.452698 + 0.477605i & 0.724632 + 0.i & -0.203482 + 0.021487i \\ 0.0746037 + 0.313061i & 0.0936111 + 0.271501i & 0.902192 + 0.i \\ 0.680724 + 0.i & 0.346776 + 0.521707i & -0.249272 + 0.286437i \end{bmatrix}$$

and

$$\Theta = \begin{bmatrix} e^{2.58364i} & 0 & 0 \\ 0 & e^{1.68825i} & 0 \\ 0 & 0 & e^{0.496512i} \end{bmatrix}.$$

Therefore,

$$H = \begin{bmatrix} 2.0261i & -0.135698 + 0.322419i & -0.228031 - 0.343709i \\ 0.135698 + 0.322419i & 0.810972i & -0.498786 + 0.313484i \\ 0.228031 - 0.343709i & 0.498786 + 0.313484i & 1.93134i \end{bmatrix},$$

and a calculation shows that $\exp(H) = U$, as desired. \square

Lemma 1.5.8. *Suppose $A, B \in \text{Mat}_{d \times d}(\mathbb{C})$. Then*

$$\exp(ABA^{-1}) = A \exp(B)A^{-1}.$$

Proof. By definition,

$$\begin{aligned} \exp(ABA^{-1}) &= I + ABA^{-1} + \frac{1}{2!}(ABA^{-1})^2 + \dots = I + ABA^{-1} + \frac{1}{2!}AB^2A^{-1} + \dots \\ &= A \left(I + B + \frac{1}{2!}B^2 + \dots \right) A^{-1} = A \exp(B)A^{-1}. \end{aligned}$$

\square

To recap, we have a surjective map $\exp: T_I \text{U}(d) \rightarrow \text{U}(d)$. Moreover, the non-injectivity of \exp is due to the periodicity of the complex exponential: $e^{i\theta} = e^{i(\theta+2\pi)}$. So \exp maps the neighborhood of the origin in $T_I \text{U}(d)$ consisting of skew-Hermitian matrices with spectral norm $< \pi$ bijectively onto the neighborhood of I in $\text{U}(d)$ consisting of unitary matrices whose eigenvalues all have argument strictly between $-\pi$ and π ; in other words, this only excludes those unitary matrices with i as an eigenvalue, which is some positive codimension subset of $\text{U}(d)$. So then \exp provides a local coordinate chart in the complement of this subset.

1.6 The Lie Bracket on Vector Fields

As we've defined it, $\mathfrak{X}(M)$ is a $C^\infty(M)$ -module:⁶ we can certainly add two vector fields or multiply a vector field by a smooth function and get a new vector field.

⁶If you haven't seen modules before, they're basically like vector fields where the ring of scalars isn't necessarily a field. In the case of $\mathfrak{X}(M)$, the smooth functions $C^\infty(M)$ serve as the scalars, but $C^\infty(M)$ is only a ring, not a field: only nowhere-vanishing functions have well-defined inverses.

In fact, there is even more algebraic structure on $\mathfrak{X}(M)$: it is something called a *Lie algebra*, which means it is a (real) vector space which admits a binary operation (called a *Lie bracket*) satisfying certain axioms.⁷

In order to define the Lie bracket of vector fields, it is instructive to think about how we could possibly define a binary operation on vector fields. For example, we could try to generalize binary operations on vector fields we already know in certain special cases. For example:

- Since $T\mathbb{R} \cong \mathbb{R} \times \mathbb{R}$, we can interpret a vector field on \mathbb{R} as a real-valued function by just recording the second entry: $X(p) = (p, v)$, and the corresponding function is $p \mapsto v$. Since functions form an algebra by pointwise multiplication, this defines a binary operation on vector fields on \mathbb{R} .
- More generally, every smooth 1-manifold has a trivial tangent bundle, so we can treat any vector field on any 1-manifold as a smooth, real-valued function and get an algebra structure on vector fields.⁸
- Since $T\mathbb{R}^2 \cong \mathbb{R}^2 \times \mathbb{R}^2$ and since we can define an equivalence $\mathbb{R}^2 \leftrightarrow \mathbb{C}$ by $(x, y) \leftrightarrow x + iy$, we can interpret a vector field on \mathbb{R}^2 (or, more generally, any surface with trivial tangent bundle) as a complex-valued function. Again, functions form an algebra by pointwise multiplication, so this defines a binary operation on vector fields on \mathbb{R}^2 .⁹
- Since $T\mathbb{R}^3 \cong \mathbb{R}^3 \times \mathbb{R}^3$, we can interpret a vector field on \mathbb{R}^3 as a vector-valued function on \mathbb{R}^3 . Then the (pointwise) cross product \times gives a binary operation on vector fields on \mathbb{R}^3 .
- Since $T\mathbb{R}^4 \cong \mathbb{R}^4 \times \mathbb{R}^4$ and since we can define an equivalence $\mathbb{R}^4 \leftrightarrow \mathbb{H}$ between \mathbb{R}^4 and the quaternions by $(t, x, y, z) \leftrightarrow t + ix + jy + kz$, we can interpret vector fields on \mathbb{R}^4 as quaternion-valued functions. Again, functions form an algebra by pointwise multiplication, so this defines a binary operation on vector fields on \mathbb{R}^4 .

This is in some sense a generalization of the cross product on \mathbb{R}^3 : if we represent $(x, y, z), (u, v, w) \in \mathbb{R}^3$ by purely imaginary quaternions $p = ix + jy + kz$ and $q = iu + jv + kw$, then the quaternion product

$$pq = -p \cdot q + p \times q,$$

where $p \cdot q$ is the dot product of the two vectors in \mathbb{R}^3 , and $p \times q$ is the cross product (interpreted as a purely imaginary quaternion).¹⁰

So the natural question is: can we generalize these examples to higher-dimensional Euclidean spaces, and more generally to arbitrary manifolds?

Unfortunately, the answer is basically “no.” As you might have heard, there are no normed division rings over \mathbb{R} besides \mathbb{R} , \mathbb{C} , and \mathbb{H} , and there aren’t even normed division algebras (where we allow multiplication to be non-associative) aside from these examples and the octonions \mathbb{O} , which are 8-dimensional.

Moreover, even if there were more normed division algebras, the operations above depended very strongly on the triviality of the tangent bundle, which is not going to work in more general manifolds.

⁷Pronunciation note: The Norwegian surname “Lie” is typically pronounced the same as the common English or Korean surname “Lee,” rather than like the English word “lie.”

⁸This correspondence between sections of trivial line bundles and functions gives you some reason to believe that the appropriate notion of “function” in algebraic geometry is often a section of a line bundle.

⁹Interpreting vector fields on the plane as complex-valued functions turns out to be quite useful in 2-dimensional fluid mechanics; look up *stream function* and *complex potential* for more.

¹⁰Look up *Clifford algebras* for a vast generalization of this kind of product operation that splits into a scalar part (the $-p \cdot q$) and a vector part (the $p \times q$).

So here's a different approach: recall that we can interpret vector fields on M as operators on $C^\infty(M)$. That is, for $f \in C^\infty(M)$, $Xf \in C^\infty(M)$ as well; this is basically the function recording the directional derivative of f in the direction of X at each point on the manifold.

An obvious thing to guess is that, if $X, Y \in \mathfrak{X}(M)$, then the composition $X \circ Y = XY$ is also a vector field. After all, $(XY)f = X(Yf)$ will be a smooth function, so XY is also an operator on $C^\infty(M)$.

Let's work in local coordinates at a point $p \in M$ to see what this operator looks like when applied to a function f which is differentiable at p .

We know that, in terms of the local coordinate basis $\left\{\frac{\partial}{\partial x_1}, \dots, \frac{\partial}{\partial x_n}\right\}$, we can write

$$X(p) = \sum_{i=1}^n a_i(p) \frac{\partial}{\partial x_i} \quad \text{and} \quad Y(p) = \sum_{i=1}^n b_i(p) \frac{\partial}{\partial x_i},$$

where the a_i and b_i are smooth functions defined in some neighborhood of p . So then

$$(XY)f = X(Yf) = X\left(\sum_{i=1}^n b_i \frac{\partial f}{\partial x_i}\right) = \sum_{i,j=1}^n a_j \frac{\partial}{\partial x_j} \left(b_i \frac{\partial f}{\partial x_i}\right) = \sum_{i,j=1}^n a_j \left(\frac{\partial b_i}{\partial x_j} \frac{\partial f}{\partial x_i} + b_i \frac{\partial^2 f}{\partial x_j \partial x_i}\right).$$

This should give you pause because it is no longer a first-order differential operator. Let's make this more obvious by rewriting as

$$(XY)f = \left[\sum_{j=1}^n \left(a_j \left(\sum_{i=1}^n \frac{\partial b_i}{\partial x_j} \frac{\partial}{\partial x_i} \right) \right) + \sum_{j=1}^n \left(a_j \left(\sum_{i=1}^n b_i \frac{\partial^2}{\partial x_j \partial x_i} \right) \right) \right] f, \quad (1.1)$$

so the stuff inside the brackets (which is XY) is a differential operator on $C^\infty(M)$, but it's not a vector field; for example, it cannot be written in terms of the local coordinate basis $\left\{\frac{\partial}{\partial x_1}, \dots, \frac{\partial}{\partial x_n}\right\}$.

Example 1.6.1. If $X = \frac{\partial}{\partial x}$ and $Y = \frac{\partial}{\partial y}$ are the standard coordinate vector fields on \mathbb{R}^2 , then (1.1) reduces to

$$(XY)f = \frac{\partial^2 f}{\partial x \partial y},$$

as you would expect from computing

$$X(Yf) = \frac{\partial}{\partial x} \left(\frac{\partial f}{\partial y} \right) = \frac{\partial^2 f}{\partial x \partial y}.$$

So $XY = \frac{\partial^2}{\partial x \partial y}$. But there's no sensible way to interpret this second-order operator as a vector field.

Example 1.6.2. If $X = r \frac{\partial}{\partial r}$ and $Y = r \frac{\partial}{\partial \theta}$ are the (scaled) radial and rotational fields, then

$$(XY)f = X(Yf) = r \frac{\partial}{\partial r} \left(r \frac{\partial f}{\partial \theta} \right) = r \frac{\partial f}{\partial \theta} + r^2 \frac{\partial^2 f}{\partial r \partial \theta},$$

so

$$XY = r \frac{\partial}{\partial \theta} + r^2 \frac{\partial}{\partial r \partial \theta}.$$

Again, a second-order operator appears.

The problem with (1.1) is the second term, which is a second-order operator. Now notice that if we had done this in the reverse order we would have gotten

$$(YX)f = \left[\sum_{j=1}^n \left(b_j \left(\sum_{i=1}^n \frac{\partial a_i}{\partial x_j} \frac{\partial}{\partial x_i} \right) \right) + \sum_{j=1}^n \left(b_j \left(\sum_{i=1}^n a_i \frac{\partial^2}{\partial x_j \partial x_i} \right) \right) \right] f. \quad (1.2)$$

This doesn't just have the same type of problem as before, it has literally the same problem: because mixed partials commute, the second terms in (1.1) and (1.2) agree. So, by subtracting, we can cancel them and get

$$(XY - YX)f = \left[\sum_{i,j=1}^n \left(\left(a_j \frac{\partial b_i}{\partial x_j} - b_j \frac{\partial a_i}{\partial x_j} \right) \frac{\partial}{\partial x_i} \right) \right] f, \quad (1.3)$$

which is written in terms of the $\left\{ \frac{\partial}{\partial x_1}, \dots, \frac{\partial}{\partial x_n} \right\}$ basis. So this really is a vector field.

Definition 1.6.3. If $X, Y \in \mathfrak{X}(M)$, the *Lie bracket* of X and Y is a vector field $[X, Y]$ defined by

$$[X, Y]f := X(Yf) - Y(Xf).$$

The key feature of this operation is that it satisfies the axioms of a Lie bracket ((i)–(iii) in Proposition 1.6.4), and hence makes $\mathfrak{X}(M)$ into a *Lie algebra*.

Proposition 1.6.4. If $X, Y, Z \in \mathfrak{X}(M)$ and $a, b \in \mathbb{R}$, $f, g \in C^\infty(M)$, then

- (i) $[X, Y] = -[Y, X]$ (anti-commutativity)
- (ii) $[aX + bY, Z] = a[X, Z] + b[Y, Z]$ (linearity)
- (iii) $[[X, Y], Z] + [[Y, Z], X] + [[Z, X], Y] = 0$ (Jacobi identity)
- (iv) $[fX, gY] = fg[X, Y] + f(Xg)Y - g(Yf)X$.

Proof. Homework. □

Example 1.6.5. Continuing with Example 1.6.2, we already computed XY , so we can also compute

$$YX = r \frac{\partial}{\partial \theta} \left(r \frac{\partial}{\partial r} \right) = r^2 \frac{\partial^2}{\partial \theta \partial r},$$

and hence

$$[X, Y] = r \frac{\partial}{\partial \theta}.$$

Example 1.6.6. Consider $M = S^2$ and the vector fields $X = \frac{\partial}{\partial \theta}$ and $Y = \frac{\partial}{\partial z}$ where (θ, z) are cylindrical coordinates on S^2 ; see Figure 1.11.

In other words, we have the local coordinate chart $\phi: (0, 2\pi) \times (-1, 1) \rightarrow S^2$ given by

$$\phi(\theta, z) = (\sqrt{1 - z^2} \cos \theta, \sqrt{1 - z^2} \sin \theta, z),$$

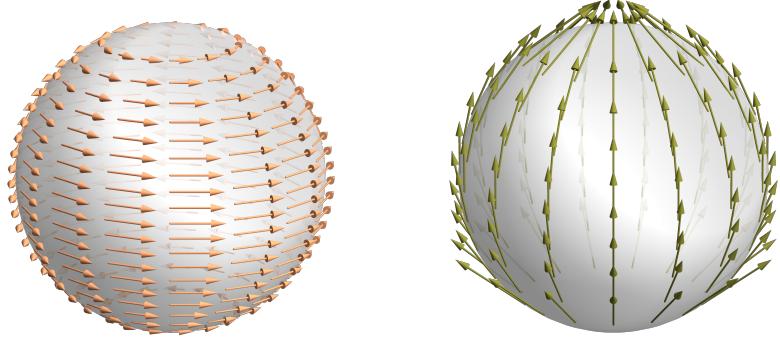


Figure 1.11: The vector fields $X = \frac{\partial}{\partial\theta}$ and $Y = \frac{\partial}{\partial z}$ on S^2 .

and X and Y are the corresponding coordinate fields.

Notice that

$$[X, Y]f = \frac{\partial^2 f}{\partial\theta\partial z} - \frac{\partial^2 f}{\partial z\partial\theta} = 0$$

since mixed partials commute.

More generally, whenever $X = \frac{\partial}{\partial x_i}$ and $Y = \frac{\partial}{\partial x_j}$ are coordinate fields in a neighborhood of a point in a manifold, $[X, Y] = 0$.

In Cartesian coordinates

$$X = \sqrt{1-z^2} \left(-\sin\theta \frac{\partial}{\partial x} + \cos\theta \frac{\partial}{\partial y} \right) = \sqrt{1-z^2} \left(-y \frac{\partial}{\partial x} + x \frac{\partial}{\partial y} \right),$$

and

$$Y = -\frac{z \cos\theta}{\sqrt{1-z^2}} \frac{\partial}{\partial x} - \frac{z \sin\theta}{\sqrt{1-z^2}} \frac{\partial}{\partial y} + \frac{\partial}{\partial z} = -\frac{xz}{1-z^2} \frac{\partial}{\partial x} - \frac{yz}{1-z^2} \frac{\partial}{\partial y} + \frac{\partial}{\partial z}.$$

So then some much more unpleasant calculations using (1.3) shows you that $[X, Y] = 0$ in these coordinates as well.

Now we look at some examples of Lie algebras not coming from vector fields.

Example 1.6.7. \mathbb{R}^3 forms a Lie algebra with the bracket operation given by the cross product: for $u, v \in \mathbb{R}^3$, define $[u, v] := u \times v$. Then it is straightforward to check that this bracket satisfies (i)–(iii) above. Written in terms of cross products, (iii) says

$$0 = (u \times v) \times w + (v \times w) \times u + (w \times u) \times v = (u \times v) \times w - u \times (v \times w) + v \times (u \times w),$$

or equivalently,

$$(u \times v) \times w = u \times (v \times w) - v \times (u \times w).$$

In particular, this records the failure of associativity of the cross product.

More generally, the Jacobi identity records the failure of associativity of the Lie bracket:

$$[[X, Y], Z] = [X, [Y, Z]] - [Y, [X, Z]].$$

Example 1.6.8. Recall [Example 1.5.3](#), in which we showed that left-invariant vector fields on $U(d)$ are of the form $X(U) = U\Delta_X$, for $\Delta_X \in T_I U(d)$, which is the collection of skew-Hermitian $d \times d$ matrices. In particular, $X(I) = \Delta_X$.

As we will see in more detail later ([Proposition 3.4.2](#)), it will turn out that the Lie bracket on left-invariant vector fields on a matrix group like $U(d)$ just corresponds to the matrix commutator operation in the tangent space to the identity.

In this case, that means that, if $X, Y \in \mathfrak{X}(U(d))$ are left-invariant, then $[X, Y]$ is also left-invariant and, for each $U \in U(d)$,

$$[X, Y](U) = U(\Delta_X \Delta_Y - \Delta_Y \Delta_X),$$

where the products inside parentheses are just matrix products between the skew-Hermitian matrices Δ_X and Δ_Y ; that is, the term in parentheses is just the usual matrix commutator.

Exercise 1.6.9. Check that the commutator of two skew-Hermitian matrices is skew-Hermitian.

This all tells you that the correspondence between left-invariant vector fields and elements of $T_I U(d)$ turns the Lie bracket of left-invariant vector fields into the matrix commutator in $T_I U(d)$. These are two different realizations of the same Lie algebra, usually called $\mathfrak{u}(d)$.

Example 1.6.10. If we play the same game with $SO(d)$, it turns out that the tangent space to the identity consists of $d \times d$ skew-symmetric matrices, and the Lie algebra on left-invariant vector fields on $SO(d)$ corresponds to the matrix commutator on skew-symmetric matrices; we'll write the collection of skew-symmetric $d \times d$ matrices as $\mathfrak{so}(d)$.

Consider the case $d = 3$. Then we can write elements of $\mathfrak{so}(3)$ as

$$\Delta = \begin{bmatrix} 0 & -z & y \\ z & 0 & -x \\ -y & x & 0 \end{bmatrix}.$$

(The reason for the funny ordering and sign choices will shortly become apparent.)

Now, $\mathfrak{so}(3)$ is a 3-dimensional vector space, and we have a vector space isomorphism $F: \mathbb{R}^3 \rightarrow \mathfrak{so}(3)$ given by

$$F: (x, y, z) \mapsto \begin{bmatrix} 0 & -z & y \\ z & 0 & -x \\ -y & x & 0 \end{bmatrix}. \quad (1.4)$$

Let $\Delta_1, \Delta_2 \in \mathfrak{so}(3)$ with

$$\Delta_i = \begin{bmatrix} 0 & -z_i & y_i \\ z_i & 0 & -x_i \\ -y_i & x_i & 0 \end{bmatrix}.$$

Then

$$[\Delta_1, \Delta_2] = \Delta_1 \Delta_2 - \Delta_2 \Delta_1 = \begin{bmatrix} 0 & x_2 y_1 - x_1 y_2 & x_2 z_1 - x_1 z_2 \\ x_1 y_2 - x_2 y_1 & 0 & y_2 z_1 - y_1 z_2 \\ x_1 z_2 - x_2 z_1 & y_1 z_2 - y_2 z_1 & 0 \end{bmatrix}.$$

If you stare at this, you might recognize the entries as being the coordinates of the cross product of the corresponding vectors:

$$\begin{bmatrix} x_1 \\ y_1 \\ z_1 \end{bmatrix} \times \begin{bmatrix} x_2 \\ y_2 \\ z_2 \end{bmatrix} = \begin{bmatrix} y_1 z_2 - y_2 z_1 \\ x_2 z_1 - x_1 z_2 \\ x_1 y_2 - x_2 y_1 \end{bmatrix}.$$

In other words, $F(v_1 \times v_2) = [F(v_1), F(v_2)]$, so F is a Lie algebra homomorphism. Since it's also a bijective linear map, it's a Lie algebra isomorphism, so we've just proved that $(\mathbb{R}^3, \times) \cong \mathfrak{so}(3)$ as Lie algebras.

Thus, the Lie bracket on vector fields on manifolds is some sort of vast generalization of the cross product on \mathbb{R}^3 . In this interpretation, $\mathbb{R}^3 \cong \mathfrak{so}(3)$ is the collection of infinitesimal rotations of 3-space, where $v \in \mathbb{R}^3$ corresponds to an infinitesimal rotation around the axis spanned by v , and the correspondence between cross products and commutators reflects the fact that, for very small $\epsilon > 0$ and unit vectors u and v , the composition of an ϵ -rotation around u with an ϵ -rotation around v is, to first order, a rotation around $u + v + \frac{\epsilon}{2}u \times v$; see the Baker–Campbell–Hausdorff formula [16, Chapter 5].

Exercise 1.6.11. Convince yourself that, for $v \in \mathbb{R}^3$ a unit vector and F as defined in (1.4), $\exp(F(\theta v))$ gives the one-parameter subgroup of rotations by angle θ around the axis spanned by v . (A full proof is kind of annoying to write down, but at least convince yourself this is true when v is a coordinate vector.)

1.7 Integral Curves and Lie Derivatives

To motivate the definition of the Lie bracket, I claimed that there was additional algebraic structure on $\mathfrak{X}(M)$ in the form of a binary operation, and then (hopefully!) convinced you that there was really only one sensible way to define such an operation.

But it's probably not at all obvious why one should have guessed that there was a binary operation on vector fields in the first place, and, though I don't actually know the history, my guess is that this was not the original motivation for the Lie bracket.

In some sense the more basic notion is that of the *Lie derivative*, which is a sort of directional derivative of vector fields which, as we will see, actually generalizes to arbitrary tensor fields.

The idea is that, given two vector fields X and Y on a manifold M , we might like to define a “directional derivative of Y in the direction of X ” operator, which we will denote $\mathcal{L}_X Y$, where the \mathcal{L} is for “Lie derivative.” It will turn out that $\mathcal{L}_X Y = [X, Y]$, which might be surprising, but if you look back to the local coordinate expression (1.3) for $[X, Y]$, you'll notice it involved differentiating the coefficients of Y with respect to X (and also X with respect to Y , but we'll shortly see why there's this symmetry).

It's worth thinking for yourself about how you might go about defining such a derivative operator, so I would encourage you to do so before reading on.

Here's a way of making this “directional derivative for vector fields” operation more precise: say we have our vector fields X and Y and we want to compute $\mathcal{L}_X Y$ at a point $p \in M$. Presumably the idea is to measure how much Y is changing as we vary p in the direction of X .

In other words, we want to integrate X to get a curve $\alpha : (-\epsilon, \epsilon) \rightarrow M$ so that $\alpha(0) = p$ and $\alpha'(t) = X(\alpha(t))$ for all $t \in (-\epsilon, \epsilon)$, then look at something like

$$\lim_{t \rightarrow 0} \frac{Y(\alpha(t)) - Y(\alpha(0))}{t}.$$

See Figure 1.12.



Figure 1.12: An attempt to differentiate Y in the direction of X .

In other words, we look at the rate of change of Y as we flow in the direction of X . Unfortunately, as written the above difference quotient doesn't make sense, but first let's talk about flows and integrating vector fields.

Definition 1.7.1. Let $X \in \mathfrak{X}(M)$. A curve $\alpha: (a, b) \rightarrow M$ is called an *integral curve* (or *trajectory*) for X if $\alpha'(t) = X(\alpha(t))$ for all $t \in (a, b)$.

Proposition 1.7.2. Let $X \in \mathfrak{X}(M)$ and let $p \in M$. Then there exists a neighborhood U of p , a $\delta > 0$, and a smooth map $\Phi: (-\delta, \delta) \times U \rightarrow M$ so that, for each $q \in U$, the map $t \mapsto \Phi(t, q)$ is the unique smooth curve satisfying the ODE $\frac{\partial \Phi}{\partial t} = X(\Phi(t, q))$ with initial condition $\Phi(0, q) = q$. See [Figure 1.13](#).



Figure 1.13: The local flow generates an integral curve through each point.

Proof. Since M is locally diffeomorphic to \mathbb{R}^n (via the coordinate charts), this follows from the fundamental theorem on existence and uniqueness of solutions to ODEs in \mathbb{R}^n . \square

We often use the notation $\Phi_t(q) := \Phi(t, q)$ and call Φ_t the *local flow* of X .

Example 1.7.3. Let $M = S^2$ and consider the vector field $X(x, y, z) = zx \frac{\partial}{\partial x} + zy \frac{\partial}{\partial y} + (z^2 - 1) \frac{\partial}{\partial z}$ (of course, this is written in extrinsic \mathbb{R}^3 coordinates rather than intrinsic coordinates). See [Figure 1.14](#).

In fact, you can check that this is just $(d\phi_N)Y$, where Y is the vector field on \mathbb{R}^2 given by $Y(u, v) = -u \frac{\partial}{\partial u} - v \frac{\partial}{\partial v}$.

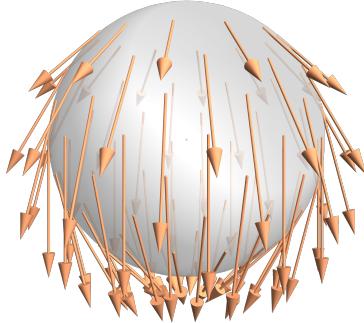


Figure 1.14: The vector field $X(x, y, z) = zx \frac{\partial}{\partial x} + zy \frac{\partial}{\partial y} + (z^2 - 1) \frac{\partial}{\partial z}$ on the sphere.

Visually, the local flow pushes the mass of the sphere down towards the south pole. We can integrate the flow explicitly in *Mathematica* to get

$$\Phi_t(x, y, z) = \frac{1}{1 + 2e^{2t} + (1 - e^{2t})z} \left(2e^t x, 2e^t y, 1 - e^{2t} + (1 + e^{2t})z \right).$$

See [Figure 1.15](#).

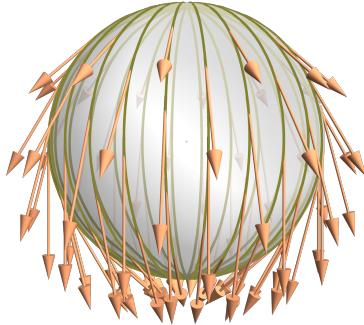


Figure 1.15: The local flow of X .

This is an example of a *complete vector field* since the local flow exists for all time at all points. If we had removed the Antarctic Circle from the sphere, this vector field would be incomplete (since the flow ceases to exist once you go over the edge).

With local flows in our toolkit we would then like to define the Lie derivative as something like

$$(\mathcal{L}_X Y)(p) = \lim_{t \rightarrow 0} \frac{Y(\Phi_t(p)) - Y(p)}{t}.$$

But if you think about it, the numerator in the difference quotient *makes no sense!* After all, $Y(\Phi_t(p)) \in T_{\Phi_t(p)}M$ and $Y(p) \in T_pM$ live in completely different vector spaces. There's no sense in which we can subtract these two vectors.

So we need some way to get the tangent vectors determined by Y at different points to be in the same tangent space.

Thus far, the only machines we have for moving tangent vectors between different tangent spaces are differentials of smooth maps. The only smooth map we have at our disposal is Φ_t , so somehow that must come into play. One possibility is

$$\lim_{t \rightarrow 0} \frac{Y(\Phi_t(p)) - (d\Phi_t)_p Y(p)}{t}$$

since both $Y(\Phi_t(p))$ and $(d\Phi_t)_p Y(p)$ live in $T_{\Phi_t(p)}M$. The problem with this is that, as t changes, $T_{\Phi_t(p)}M$ is also changing, so we're taking limits of vectors in different tangent spaces. Now, all these tangent vectors still live in TM , so this is in fact doable, but it would be better if all the vectors in the limit actually lived in the tangent space, ideally T_pM .

We are finally ready to define the Lie derivative:

Definition 1.7.4. Suppose M is a manifold, $X, Y \in \mathfrak{X}(M)$, and $p \in M$. Then the *Lie derivative* of Y with respect to X at p is

$$(\mathcal{L}_X Y)(p) := \lim_{t \rightarrow 0} \frac{(d\Phi_{-t})_{\Phi_t(p)} Y(\Phi_t(p)) - Y(p)}{t},$$

where Φ_t is the local flow for X .

This is a lot of obnoxious notation, but let's try to unpack it. First of all, notice that $\Phi_{-t}(\Phi_t(p)) = p$, since we're just flowing forward in time by t and then backwards for the same time. So

$$(d\Phi_{-t})_{\Phi_t(p)} : T_{\Phi_t(p)}M \rightarrow T_pM.$$

Hence, plugging in $Y(\Phi_t(p))$ gives $(d\Phi_{-t})_{\Phi_t(p)} Y(\Phi_t(p)) \in T_pM$, and the difference quotient and the limit make sense as happening entirely in T_pM . A (somewhat messy) picture is shown in [Figure 1.16](#).



Figure 1.16: The pieces in the definition of $\mathcal{L}_X Y$.

In looking at [Figure 1.16](#), recall that the point was that we wanted to see how Y varied as we moved in the direction of X . So at a point $p \in M$, we follow the local flow of X for a small time t . Y determines a tangent vector at the resulting point $\Phi_t(p)$. Now we push this vector $Y(\Phi_t(p))$ to T_pM by the negative local flow of X and compare to $Y(p)$.

You can imagine that if Y had a constant magnitude and made a constant angle with X , then pushing Y forward by the negative local flow (which is exactly $(d\Phi_{-t})_{\Phi_t(p)}Y(\Phi_t(p))$) would just give you the same vector you started with, namely $Y(p)$. But this makes sense: if Y has a constant magnitude and makes a constant angle with X , then it's not changing at all with respect to X , and this derivative should really be 0.¹¹

The slightly amazing fact is that the Lie derivative and the Lie bracket are the same:

Proposition 1.7.5. *Let $X, Y \in \mathfrak{X}(M)$ and choose $p \in M$. Then*

$$[X, Y](p) = (\mathcal{L}_X Y)(p).$$

Notice, in particular, that this shows that the Lie derivative is antisymmetric:

$$\mathcal{L}_Y X = [Y, X] = -[X, Y] = -\mathcal{L}_X Y$$

by the antisymmetry of the Lie bracket, which you might not have guessed from the definition of the Lie derivative.

Proof of Proposition 1.7.5. Both expressions are tangent vectors at p , so formally they're differential operators evaluated on functions that are differentiable in a neighborhood of p . So we'll choose a test function f which is differentiable in a neighborhood of p , and the goal is to show that

$$\lim_{t \rightarrow 0} \frac{(d\Phi_{-t})_{\Phi_t(p)}Y(\Phi_t(p)) - Y(p)}{t}(f)(p) = ((XY - YX)f)(p).$$

Let's focus on the first term in the numerator, $((d\Phi_{-t})_{\Phi_t(p)}Y)f(p)$:

$$((d\Phi_{-t})_{\Phi_t(p)}Y)f(p) = (df_p)((d\Phi_{-t})_{\Phi_t(p)}Y(p)) = (d(f \circ \Phi_{-t}))_{\Phi_t(p)}Y(\Phi_t(p)) = (Y(f \circ \Phi_{-t}))(\Phi_t(p)),$$

where we used Lemma 1.3.4 for the first and last equalities¹² and the chain rule for the middle equality. Therefore,

$$(\mathcal{L}_X Y)(p) = \lim_{t \rightarrow 0} \frac{(Y(f \circ \Phi_{-t}))(\Phi_t(p)) - (Yf)(p)}{t}.$$

If we could turn this into something like

$$\lim_{t \rightarrow 0} \frac{(Yf)(\Phi_t(p)) - (Yf)(p)}{t} = \frac{d((Yf) \circ \Phi_t)}{dt} \Big|_{t=0} = (X(Yf))(p), \quad (1.5)$$

we'd be in business.

In fact, if we could write $f(\Phi_{-t}(q)) = f(q) + tg(t, q)$ for some function g , then we could get a term

¹¹You should take this paragraph figuratively and not literally: we don't have any notion of magnitude or angle for tangent vectors, because we don't yet have inner products on the tangent spaces. That will come when we start talking about Riemannian metrics!

¹²Note: I added this lemma after the initial draft, so it was not there when you previously read that section.

like (1.5) to appear because then we'd have

$$\begin{aligned}
\lim_{t \rightarrow 0} \frac{(Y(f \circ \Phi_{-t}))(\Phi_t(p)) - (Yf)(p)}{t} &= \lim_{t \rightarrow 0} \frac{(Y(f + tg(t, \cdot)))(\Phi_t(p)) - (Yf)(p)}{t} \\
&= \lim_{t \rightarrow 0} \frac{(Yf)(\Phi_t(p)) + (Y(tg(t, \cdot)))(\Phi_t(p)) - (Yf)(p)}{t} \\
&= \lim_{t \rightarrow 0} \frac{(Yf)(\Phi_t(p)) + t(Y(g(t, \cdot)))(\Phi_t(p)) - (Yf)(p)}{t} \\
&= \lim_{t \rightarrow 0} \frac{(Yf)(\Phi_t(p)) - (Yf)(p)}{t} + Y(g(0, \Phi_0(p))) \\
&= (X(Yf))(p) + Y(g(0, p)).
\end{aligned}$$

This will then equal $((XY - YX)f)(p)$ if we can find such a g so that the expression $Y(g(0, p))$ is equal to $(Y(Xf))(p)$.

But this is easy: rearranging $f(\Phi_{-t}(q)) = f(q) + tg(t, q)$ yields

$$g(t, q) = \frac{f(\Phi_{-t}(q)) - f(q)}{t},$$

at least for $t \neq 0$. We can extend the definition to $t = 0$ by taking the limit:

$$g(0, q) := \lim_{t \rightarrow 0} \frac{f(\Phi_{-t}(q)) - f(q)}{t} = (Xf)(q),$$

which is exactly what we need, since then $Y(g(0, p)) = (Y(Xf))(p)$. \square

1.8 Extended Example on S^3

Let S^3 be the unit sphere in $\mathbb{R}^4 \cong \mathbb{C}^2 \cong \mathbb{H}$, where

$$\mathbb{H} = \{a + bi + cj + dk : a, b, c, d \in \mathbb{R}\}$$

is the division ring of quaternions. We have the defining conditions

$$i^2 = j^2 = k^2 = -1, \quad ijk = -1,$$

which for example imply that

$$-ij = ijk^2 = (ijk)k = -k \Rightarrow ij = k$$

and

$$-ji = ji(ijk) = ji^2jk = -j^2k = k \Rightarrow ji = -k,$$

so \mathbb{H} is noncommutative. \mathbb{H} also has a conjugation operation defined by

$$\overline{a + bi + cj + dk} = a - bi - cj - dk$$

and an inner product given by,

$$\langle p, q \rangle = \operatorname{Re}(p\bar{q}).$$

When $p = a + bi + cj + dk$ and $q = t + xi + yj + zk$, we see that

$$\langle p, q \rangle = \operatorname{Re}(p\bar{q}) = \operatorname{Re}((a + bi + cj + dk)(t - xi - yj - zk)) = at + bx + cy + dz,$$

so this agrees with the usual dot product on \mathbb{R}^4 . Moreover, the induced norm

$$\|p\|^2 = \langle p, p \rangle = \operatorname{Re}(p\bar{p}) = \operatorname{Re}((a + bi + cj + dk)(a - bi - cj - dk)) = a^2 + b^2 + c^2 + d^2$$

is the standard one.

Thought of as the unit quaternions $\{q \in \mathbb{H} : \|q\| = 1\}$, it becomes clear that S^3 is a group: if $p = a + bi + cj + dk, q = t + xi + yj + zk \in \mathbb{H}$ are both unit quaternions, meaning that

$$a^2 + b^2 + c^2 + d^2 = p\bar{p} = \|p\|^2 = 1 = \|q\|^2 = q\bar{q} = t^2 + x^2 + y^2 + z^2,$$

then their product is also a unit quaternion:

$$\|pq\| = (pq)(\bar{pq}) = pq\bar{q}\bar{p} = p1\bar{p} = p\bar{p} = 1.$$

To give some other terminology, S^3 is the *symplectic group* $\operatorname{Sp}(1)$; that is, the quaternionic analog of the unitary group for \mathbb{H}^1 . (We will eventually see that this group is also isomorphic to $\operatorname{SU}(2)$ and $\operatorname{Spin}(3)$).

Geometrically, it is clear that, for any $p \in S^3$, the tangent space $T_p S^3$ can be thought of as the quaternions which are orthogonal to p . So for example the tangent space to the identity $1 \in S^3$ is

$$T_1 S^3 = \{xi + yj + zk : x, y, z \in \mathbb{R}^3\},$$

the purely imaginary quaternions. Using the same idea as in [Example 1.5.3](#), we can push any tangent vector at the identity around to get a left-invariant vector field on all of S^3 . In particular, we get three mutually perpendicular vector fields $X, Y, Z \in \mathfrak{X}(S^3)$ given by

$$X(p) = pi, \quad Y(p) = pj, \quad Z(p) = pk$$

for all $p \in S^3$. (Strictly speaking, $X(p) = (dL_p)_1 i$, where L_p is left-multiplication by p , and i is interpreted as an element of $T_1 S^3$ [and similarly for Y and Z], but L_p being linear implies that $(dL_p)_1 i = pi$.)

To check that, say, $X(p)$ is really a tangent vector at p , notice that

$$\langle p, X(p) \rangle = \langle p, pi \rangle = \langle a + bi + cj + dk, -b + ai + dj - ck \rangle = -ab + ab - cd + cd = 0,$$

so $X(p) \perp p$ and hence $X(p) \in T_p S^3$. Similar calculations show that $Y(p), Z(p) \in T_p S^3$.

Moreover X, Y , and Z are mutually perpendicular everywhere: for example

$$\langle X(p), Y(p) \rangle = \langle pi, pj \rangle = \langle -b + ai + dj - ck, -c - di + aj + bk \rangle = bc - ad + ad - bc = 0.$$

(A better argument: multiplication by a unit quaternion is an isometry of $\mathbb{H} = \mathbb{R}^4$, so orthogonality of $X(1) = i$ and $Y(1) = j$ carries over to $T_p S^3$.)

Now, the corresponding local flows are

$$\xi_t(p) = p(\cos t + i \sin t), \quad \psi_t(p) = p(\cos t + j \sin t), \quad \zeta_t(p) = p(\cos t + k \sin t),$$

as we can see by differentiating; for example,

$$\frac{d}{dt} \Big|_{t=0} \xi_t(p) = \frac{d}{dt} \Big|_{t=0} p(\cos t + i \sin t) = -p \sin(0) + pi \cos(0) = pi = X(p).$$

Now we compute some Lie brackets. For example,

$$\begin{aligned} [X, Y](p) &= (\mathcal{L}_X Y)(p) = \lim_{t \rightarrow 0} \frac{(d\xi_{-t})_{\xi_t(p)} Y(\xi_t(p)) - Y(p)}{t} \\ &= \frac{d}{dt} \Big|_{t=0} (d\xi_{-t})(Y(\xi_t(p))) \\ &= \frac{d}{dt} \Big|_{t=0} (d\xi_{-t})(\xi_t(p)j) \\ &= \frac{d}{dt} \Big|_{t=0} (d\xi_{-t})(p(\cos t + i \sin t)j) \\ &= \frac{d}{dt} \Big|_{t=0} (d\xi_{-t})(p(j \cos t + k \sin t)). \end{aligned}$$

The differential $(d\xi_{-t})_{\xi_t(p)} : T_{\xi_t(p)} S^3 \rightarrow T_p S^3$, but we can interpret both of these as subspaces of the ambient space \mathbb{R}^4 , so the differential can be represented by a 4×4 matrix, namely

$$(d\xi_{-t})_{\xi_t(p)} = \begin{bmatrix} \cos t & \sin t & 0 & 0 \\ -\sin t & \cos t & 0 & 0 \\ 0 & 0 & \cos t & -\sin t \\ 0 & 0 & \sin t & \cos t \end{bmatrix},$$

so

$$(d\xi_{-t})(p(j \cos t + k \sin t)) = \begin{bmatrix} \cos t & \sin t & 0 & 0 \\ -\sin t & \cos t & 0 & 0 \\ 0 & 0 & \cos t & -\sin t \\ 0 & 0 & \sin t & \cos t \end{bmatrix} \begin{bmatrix} -c \cos t - d \sin t \\ c \sin t - d \cos t \\ a \cos t - b \sin t \\ a \sin t + b \cos t \end{bmatrix} = \begin{bmatrix} -c \cos 2t - d \sin 2t \\ c \sin 2t - d \cos 2t \\ a \cos 2t - b \sin 2t \\ a \sin 2t + b \cos 2t \end{bmatrix}.$$

Putting this all together, then

$$[X, Y](p) = \frac{d}{dt} \Big|_{t=0} (d\xi_{-t})(p(j \cos t + k \sin t)) = \begin{bmatrix} -2d \\ 2c \\ -2b \\ 2a \end{bmatrix} = 2pk = 2Z(p).$$

In other words, $[X, Y] = 2Z$ and, by similar arguments, $[Y, Z] = 2X$ and $[Z, X] = 2Y$.

In our language from [Example 1.5.3](#), X , Y , and Z are left-invariant vector fields on the Lie group S^3 , and so you should expect that the Lie bracket on left-invariant vector fields on S^3 corresponds to a Lie algebra structure on $T_1 S^3$, which is a 3-dimensional vector space. We'll call this 3-dimensional Lie algebra $\mathfrak{sp}(1)$ since $S^3 = \text{Sp}(1)$.

Thinking of S^3 and $\mathfrak{sp}(1)$ as living in the space of 1×1 quaternionic matrices (that is, inside \mathbb{H}), we might expect that the matrix commutator on $\mathfrak{sp}(1) = \{xi + yj + zk\}$ should correspond to the above Lie

bracket. Indeed, the commutator of the 1×1 matrices i and j is

$$ij - ji = k - (-k) = 2k,$$

which agrees with the above calculation that $[X, Y] = 2Z$.

Since the only Lie algebra structure we've seen on a 3-dimensional vector space is that of (\mathbb{R}^3, \times) , or equivalently $(\mathfrak{so}(3), [\cdot, \cdot])$, you might guess that $\mathfrak{sp}(1)$ is just another iteration of the same Lie algebra.

Indeed, if e_1, e_2, e_3 is the standard basis for \mathbb{R}^3 and we define $F: \mathbb{R}^3 \rightarrow \mathfrak{sp}(1)$ by

$$F(e_1) = \frac{1}{2}X, \quad F(e_2) = \frac{1}{2}Y, \quad F(e_3) = \frac{1}{2}Z,$$

then we see that, e.g.,

$$F(e_1 \times e_2) = F(e_3) = \frac{1}{2}Z = \frac{1}{4}[X, Y] = \left[\frac{1}{2}X, \frac{1}{2}Y \right] = [F(e_1), F(e_2)],$$

and more generally it's easy to check that $F(u \times v) = [F(u), F(v)]$ for any $u, v \in \mathbb{R}^3$, so F is an isomorphism of Lie algebras.

This also implies that $\mathfrak{so}(3)$ and $\mathfrak{sp}(1)$ are isomorphic Lie algebras, even though the Lie groups $\text{SO}(3)$ and S^3 are not even homotopy equivalent (for example, $\pi_1(\text{SO}(3)) \cong \mathbb{Z}/2\mathbb{Z}$ and $\pi_1(S^3) = \{1\}$), which is going to imply that they cannot be isomorphic as Lie groups.¹³

¹³That said, $\text{SO}(3) \cong \mathbb{RP}^3$ has S^3 as its universal cover; indeed, this is really what the isomorphism of Lie algebras implies.

Chapter 2

Differential Forms

2.1 Integration, Forms, and an Informal Definition

Now that we've defined vector fields and explored some special features (like the Lie bracket), the next objects to turn our attention to are *differential forms*. Differential forms turn out to be the right tool with which to define integration on manifolds, and they also turn out to encode the cohomology ring of a manifold, which means they have both a natural binary operation (corresponding to the cup product in cohomology) and a natural derivative operation (corresponding to the coboundary map).

To try to build up some intuition for differential forms, think back to vector calculus and computing integrals of the form

$$\int_A f \, dx_1 \dots dx_n, \tag{2.1}$$

where $A \subseteq \mathbb{R}^n$ is an open set (or maybe the closure of an open set), and $f : A \rightarrow \mathbb{R}$ is a (sufficiently nice) function.¹ If you stare at (2.1), and in particular at the expression $f \, dx_1 \dots dx_n$ being integrated, what exactly is this thing?

To get a handle on what it is, it's helpful to see how it transforms. If $B \subseteq \mathbb{R}^n$ is open and $g : B \rightarrow A$ is a diffeomorphism,² then

$$\int_A f \, dx_1 \dots dx_n = \int_B (f \circ g) |\det Jg| \, dy_1 \dots dy_n,$$

where I'm using y_1, \dots, y_n to indicate the coordinates on B and Jg is the Jacobian matrix of g . So now whatever kind of object $f \, dx_1 \dots dx_n$ is, it's the same type of object as $(f \circ g) |\det Jg| \, dy_1 \dots dy_n$, and this describes how it transforms under diffeomorphisms.

Now, examining $(f \circ g) |\det Jg| \, dy_1 \dots dy_n$, it's clear that this is a more complicated object than just a function. In particular, the presence of the determinant tells you that this is some sort of *alternating, multilinear* object. Recall the definitions:

Definition 2.1.1. Let V_1, \dots, V_k, W be R -modules. A map

$$f : V_1 \times \dots \times V_k \rightarrow W$$

¹For our purposes we'll just be dealing with the Riemann integral, but one can also generalize the Lebesgue integral to manifolds.

²In calculus terms, this is a change of variables; e.g., a transition from Cartesian to spherical coordinates.

is *multilinear* if it is linear in each factor. That is, for any $j \in \{1, \dots, k\}$, if $v_i \in V_i$ for each $i = 1, \dots, k$, $u_j \in V_j$, and $c \in \mathbb{R}$, then

$$f(v_1, \dots, u_j + cv_j, \dots, v_k) = f(v_1, \dots, u_j, \dots, v_k) + cf(v_1, \dots, v_j, \dots, v_k).$$

Definition 2.1.2. Let V and W be R -modules. A multilinear map $f: V^k \rightarrow W$ is *alternating* if, for any $i \neq j$, $v_i = v_j$ implies that $f(v_1, \dots, v_k) = 0$.

Thinking of the determinant map as a function $\det: V^n \rightarrow \mathbb{R}$ where $(v_1, \dots, v_n) \in V^n$ are interpreted as the columns of a matrix whose determinant is then computed, \det is multilinear and alternating. Indeed, these two properties uniquely characterize the determinant up to scale:

Theorem 2.1.3. Suppose V is an n -dimensional vector space. Up to scaling by a constant factor, \det is the unique alternating, multilinear map $V^n \rightarrow \mathbb{R}$.

The point of all of this is that the things we actually integrate in multivariable calculus are alternating, multilinear gadgets. Informally, this is how we should think about differential forms on manifolds:

Definition 2.1.4 (Informal Definition). A smooth *differential k -form* on a manifold M is a smooth, alternating, $C^\infty(M)$ -multilinear map

$$\omega: \mathfrak{X}(M)^k \rightarrow C^\infty(M)$$

(recall that $\mathfrak{X}(M)$ is a $C^\infty(M)$ -module). The vector space of k -forms is denoted $\Omega^k(M)$.

In other words, at each point $p \in M$, a k -form ω will input k tangent vectors at p and output a real number in an alternating, multilinear way. You might guess that this means that ω is *really* a (smooth) section of some vector bundle over M , and indeed we will shortly define it in this way.

But before that, let's look at some examples of things that should be differential forms according to whatever formal definition we eventually give.

Example 2.1.5. Let $k = 0$. What is a 0-form? Just from looking at [Definition 2.1.4](#), it's supposed to be something which inputs 0 vector fields on M and outputs a smooth function on M . But then that means it has no input and a smooth function as output, so is really just that smooth function. So $C^\infty(M) = \Omega^0(M)$.

Example 2.1.6. If $k > \dim(M)$, then the alternating condition implies that $\Omega^k(M) = \{0\}$.

Example 2.1.7. Suppose $f \in C^\infty(M)$; that is, $f: M \rightarrow \mathbb{R}$ is smooth. Recall that, for each $p \in M$, the differential $df_p: T_p M \rightarrow T_{f(p)} \mathbb{R}$. But $T_{f(p)} \mathbb{R}$ can be canonically identified with \mathbb{R} , so we can interpret df_p as a linear map $T_p M \rightarrow \mathbb{R}$.

In plainer language, df inputs a vector field on M and outputs a number at each point—that is, a function on M . So in fact, since the alternating condition is vacuous with only one input, $df \in \Omega^1(M)$.

(Foreshadowing: $f \in \Omega^0(M)$ and $df \in \Omega^1(M)$, so you might ask whether in general there's some operation d which turns elements of $\Omega^k(M)$ into elements of $\Omega^{k+1}(M)$.)

Example 2.1.8. Suppose $\alpha \in \Omega^1(M)$. Then at each point α inputs a tangent vector and outputs a number in a linear way. In other words, for each $p \in M$, $\alpha_p \in (T_p M)^*$, the dual space of $T_p M$, also called the cotangent space at p .

Just as a vector field is a (smooth) choice of tangent vector at each point, this says that a 1-form is a (smooth) choice of a cotangent vector at each point. More formally, a vector field is a section of the tangent bundle TM and so, as we'll see, a 1-form is a section of the cotangent bundle T^*M .

Example 2.1.9. Suppose $M = \mathbb{R}^n$ endowed with its usual dot product. This induces an inner product on each tangent space $T_p \mathbb{R}^n$, which we will (confusingly, but in keeping with the usual practice in Riemannian geometry) denote by g . Specifically, we define g_p to be an inner product on $T_p \mathbb{R}^n$ given by $g_p(u, v) = u \cdot v$ for any $u, v \in T_p \mathbb{R}^n$.

Now, let $X \in \mathfrak{X}(\mathbb{R}^n)$. Then X corresponds to a unique $\alpha \in \Omega^1(\mathbb{R}^n)$ defined as follows: for $Y \in \mathfrak{X}(\mathbb{R}^n)$ and $p \in \mathbb{R}^n$, $\alpha(Y) \in C^\infty(\mathbb{R}^n)$ is given by

$$(\alpha(Y))(p) := g_p(X(p), Y(p))$$

(again, this is just the dot product of the vector $X(p)$ with the vector $Y(p)$). Then α is certainly a map $\mathfrak{X}(\mathbb{R}^n) \rightarrow C^\infty(\mathbb{R}^n)$, and it's smooth because X and g_p are. It's trivially alternating, and it's linear since g_p is. So $\alpha \in \Omega^1(\mathbb{R}^n)$.

In fact, any $\beta \in \Omega^1(\mathbb{R}^n)$ can be written in this way: at each $p \in \mathbb{R}^n$, β determines a linear functional $\beta_p \in (T_p \mathbb{R}^n)^*$. But then the Riesz Representation Theorem tells us that there exists $u_p \in T_p \mathbb{R}^n$ so that $\beta_p(v) = g_p(u_p, v)$ for all $v \in T_p \mathbb{R}^n$. In turn, we can define a vector field $U \in \mathfrak{X}(\mathbb{R}^n)$ by $U(p) := u_p$, and the smoothness of β will turn out to imply the smoothness of U .

This all tells us that g determines an isomorphism $\flat: \mathfrak{X}(\mathbb{R}^n) \rightarrow \Omega^1(\mathbb{R}^n)$ given by

$$X^\flat(Y) = g(X, Y).$$

In fact, there's nothing special about \mathbb{R}^n in the above. The same holds on any manifold M when g is a choice of *Riemannian metric* on M .³ Since it will turn out that we can always put a Riemannian metric on any manifold, this will tell us that, for any M ,

$$\Omega^1(M) \cong \mathfrak{X}(M)$$

either as vector spaces or as $C^\infty(M)$ -modules. You should view this isomorphism as the differential geometry analog of the isomorphism between a Hilbert space and its dual.

Example 2.1.10. We'll define $\omega \in \Omega^2(\mathbb{R}^2)$ as follows. Since we have global coordinates on \mathbb{R}^2 , we can write any $v \in T_p \mathbb{R}^2$ as

$$v = a \frac{\partial}{\partial x} + b \frac{\partial}{\partial y}$$

where $\left\{ \frac{\partial}{\partial x}, \frac{\partial}{\partial y} \right\}$ is just the standard basis written in the style of our local coordinate bases,⁴ and so any vector field

$$X(p) = a(p) \frac{\partial}{\partial x} + b(p) \frac{\partial}{\partial y}$$

where now $a, b \in C^\infty(M)$.

If $X, Y \in \mathfrak{X}(\mathbb{R}^2)$ are given in coordinates by $X = a \frac{\partial}{\partial x} + b \frac{\partial}{\partial y}$ and $Y = c \frac{\partial}{\partial x} + d \frac{\partial}{\partial y}$, define

$$\omega(X, Y) = \omega \left(a \frac{\partial}{\partial x} + b \frac{\partial}{\partial y}, c \frac{\partial}{\partial x} + d \frac{\partial}{\partial y} \right) := ad - bc = \begin{vmatrix} a & b \\ c & d \end{vmatrix}.$$

Since I've written this as a determinant, it is obviously alternating and multilinear, though that can also be checked directly.

³More precise definition to come, but basically a smooth choice of inner product on each tangent space.

⁴Formally, $\frac{\partial}{\partial x}$ and $\frac{\partial}{\partial y}$ really depend on the base point p , but since they're all parallel, I'm omitting this dependence.

So this ω is really a 2-form on \mathbb{R}^2 , which we're going to write as $\omega = dx \wedge dy$. The way to read this notation is as follows: dx pairs with $\frac{\partial}{\partial x}$ to produce 1, and dy pairs with $\frac{\partial}{\partial y}$ to produce 1; in other words, $\{dx, dy\}$ is the dual basis to $\left\{\frac{\partial}{\partial x}, \frac{\partial}{\partial y}\right\}$. Moreover, the wedge symbol forces this to be alternating and multilinear, so that, for example

$$dx \wedge dy \left(\frac{\partial}{\partial x}, \frac{\partial}{\partial y} \right) = 1 \quad \text{but} \quad dx \wedge dy \left(\frac{\partial}{\partial y}, \frac{\partial}{\partial x} \right) = -1.$$

Notice that $dx \wedge dy$ simply returns the signed area of the quadrilateral spanned by whatever pair of vectors is fed into it.

Example 2.1.11. There was nothing special about dimension 2 in the above example. [Theorem 2.1.3](#) implies that determinants are essentially the only way to get n -forms on n -dimensional manifolds, which recall are supposed to be alternating, multilinear maps $\omega: \mathfrak{X}(M)^n \rightarrow C^\infty(M)$: at each $p \in M$, such an ω is some scalar multiple of “the determinant” (whatever that precisely means in this general setting). This scalar is allowed to vary as we move around, so it turns out that

$$\omega = f \, dVol_M,$$

where $dVol_M$ is the name we give to the n -form which is just the determinant (again, whatever that really means, and with the caveat that $dVol_M$ does not always exist) on each $T_p M$, and f is some smooth function.

While this is obviously far from a rigorous proof, this hopefully gives you some intuition to the fact that $\Omega^n(M) \cong C^\infty(M)$.⁵

Just as with the previous example, geometrically $dVol_M$ is returning the signed n -dimensional volume of the parallelepiped spanned by any n vectors fed into it. This explains the notation and the terminology: we call this form a *volume form*.

Example 2.1.12. Let \det be the determinant on \mathbb{R}^n , which I think of as an alternating, multilinear map $\det: (\mathbb{R}^n)^n \rightarrow \mathbb{R}$. In fact, I'll use the same notation to indicate the induced map $\mathfrak{X}(\mathbb{R}^n)^n \rightarrow C^\infty(\mathbb{R}^n)$ by applying the determinant at each point. As in the previous example, I can think of $\det \in \Omega^n(\mathbb{R}^n)$.

But now we're going to combine \det with a vector field to define an $(n-1)$ -form η . Specifically, suppose $X \in \mathfrak{X}(\mathbb{R}^n)$. We'll define η pointwise, so let $p \in \mathbb{R}^n$. Since η is going to be an $(n-1)$ -form, it is supposed to input $n-1$ tangent vectors at p and output a number. To do so, we'll just tack on $X(p)$ and plug into \det ; that is, for $v_1, \dots, v_{n-1} \in T_p \mathbb{R}^n$, define

$$\eta_p(v_1, \dots, v_{n-1}) := \det(X(p), v_1, \dots, v_{n-1}).$$

This is alternating and multilinear, and the smoothness of X will imply it is smooth, so $\eta \in \Omega^{n-1}(M)$. Moreover, every $(n-1)$ -form can be written in this way, so this implies that $\mathfrak{X}(\mathbb{R}^n)$ is isomorphic to $\Omega^{n-1}(\mathbb{R}^n)$.

Again, as in [Example 2.1.9](#), this turns out to work on general manifolds when we have an analog of \det , which is our volume form from [Example 2.1.11](#). In general, then, this will imply that, for an orientable n -manifold M , $\Omega^1(M) \cong \Omega^{n-1}(M)$.

⁵Strictly speaking, this argument only works when M is orientable, as we will see.

2.2 Differential Forms and Vector Calculus

Suppose we restrict to the case of a (Riemannian) 3-manifold; that is, a 3-dimensional manifold M endowed with a Riemannian metric g (again, a Riemannian metric is a [smooth] choice of inner product g_p on each tangent space; if this makes you uncomfortable, just assume $M = \mathbb{R}^3$ and we just have the standard dot product on each tangent space). This section is a bit weird and definitely very informal: I'm using a bunch of stuff that I haven't actually defined yet. But the point is to try to say that concepts like *differential forms* and *exterior derivatives* that we will eventually define carefully are just generalizations of things you already understand very well.

From Examples 2.1.5–2.1.12, we know that

$$\Omega^0(M) = C^\infty(M) \cong \Omega^3(M), \quad \Omega^1(M) \cong \mathfrak{X}(M) \cong \Omega^2(M), \quad \text{and} \quad \Omega^k(M) = \{0\} \text{ for } k > 3.$$

So anything we can say about differential forms on M must be expressible just in terms of vector fields and functions. We know from vector calculus that there are various differentiation operators on functions and vector fields, namely div , grad , and curl , so what do these mean at the level of forms?

Or, turning it around, are there operations on forms which fill in the squares in the following diagram and makes them commute?

$$\begin{array}{ccccccc} \Omega^0(M) & & \Omega^1(M) & & \Omega^2(M) & & \Omega^3(M) \\ \| & & \cong \uparrow \flat & & \cong \uparrow & & \cong \uparrow \\ C^\infty(M) & \xrightarrow{\nabla} & \mathfrak{X}(M) & \xrightarrow{\nabla \times} & \mathfrak{X}(M) & \xrightarrow{\nabla \cdot} & C^\infty(M) \end{array}$$

First, we (hopefully) recall from vector calculus that the gradient of a function is a vector field, which can be computed in local coordinates as

$$\nabla f = \sum_{i=1}^3 \frac{\partial f}{\partial x_i} \frac{\partial}{\partial x_i}.$$

Exercise 2.2.1. If it's not clear, make the effort to connect this to the gradient you're familiar with.

Then the corresponding 1-form $(\nabla f)^\flat$ is defined at $p \in M$ by

$$(\nabla f)_p^\flat(Y) = g_p(\nabla f, Y) = g_p \left(\sum_{i=1}^3 \frac{\partial f}{\partial x_i} \frac{\partial}{\partial x_i}, \sum_{i=1}^3 b_i(p) \frac{\partial}{\partial x_i} \right) = \sum_{i,j=1}^3 f_{x_i}(p) b_j(p) g_p \left(\frac{\partial}{\partial x_i}, \frac{\partial}{\partial x_j} \right),$$

where $Y = \sum_{i=1}^3 b_i \frac{\partial}{\partial x_i}$ in local coordinates and I've written f_{x_i} as a shorthand for $\frac{\partial f}{\partial x_i}$.

If we were able to arrange it so that the local coordinate basis $\left\{ \frac{\partial}{\partial x_1}, \frac{\partial}{\partial x_2}, \frac{\partial}{\partial x_3} \right\}$ were orthonormal with respect to g_p , then this would simplify as

$$(\nabla f)_p^\flat(Y) = \sum_{i=1}^3 f_{x_i}(p) b_i(p) = \sum_{i=1}^3 b_i(p) \frac{\partial f}{\partial x_i}(p) = \left(\left(\sum_{i=1}^3 b_i \frac{\partial}{\partial x_i} \right) f \right)(p) = (Yf)(p) = df_p(Y)$$

by Lemma 1.3.4. In other words, $(\nabla f)^\flat = df$.

Recall (Example 2.1.8) that at each point a 1-form corresponds to a cotangent vector, so the local coordinate basis $\left\{\frac{\partial}{\partial x_1}, \frac{\partial}{\partial x_2}, \frac{\partial}{\partial x_3}\right\}$ induces a dual basis which we call $\{dx_1, dx_2, dx_3\}$ for $(T_p M)^*$ defined by

$$dx_i \left(\frac{\partial}{\partial x_j} \right) = \delta_{ij}.$$

In these terms,

$$df = \frac{\partial f}{\partial x_1} dx_1 + \frac{\partial f}{\partial x_2} dx_2 + \frac{\partial f}{\partial x_3} dx_3.$$

In other words, in the presence of a Riemannian metric,⁶ computing the differential of a function on a 3-manifold (in fact, this works equally well on an n -manifold) corresponds to taking the gradient of the function.

Turning to curl, recall that, if $X \in \mathfrak{X}(M)$ is written in local coordinates as $X = a_1 \frac{\partial}{\partial x_1} + a_2 \frac{\partial}{\partial x_2} + a_3 \frac{\partial}{\partial x_3}$, then

$$\nabla \times X = \left(\frac{\partial a_3}{\partial x_2} - \frac{\partial a_2}{\partial x_3} \right) \frac{\partial}{\partial x_1} + \left(\frac{\partial a_1}{\partial x_3} - \frac{\partial a_3}{\partial x_1} \right) \frac{\partial}{\partial x_2} + \left(\frac{\partial a_2}{\partial x_1} - \frac{\partial a_1}{\partial x_2} \right) \frac{\partial}{\partial x_3}.$$

As discussed in Example 2.1.12, this corresponds to an element of $\omega^{3-1}(M) = \Omega^2(M)$ given by

$$\left(\frac{\partial a_3}{\partial x_2} - \frac{\partial a_2}{\partial x_3} \right) dx_2 \wedge dx_3 + \left(\frac{\partial a_1}{\partial x_3} - \frac{\partial a_3}{\partial x_1} \right) dx_3 \wedge dx_1 + \left(\frac{\partial a_2}{\partial x_1} - \frac{\partial a_1}{\partial x_2} \right) dx_1 \wedge dx_2.$$

For reasons to be explained later, I'm going to call this 2-form $\star(\nabla \times X)^\flat$.

So now what is the operation on $X^\flat = a_1 dx_1 + a_2 dx_2 + a_3 dx_3$ which would have produced $\star(\nabla \times X)^\flat$? The idea is to compute the differential (which is a 1-form) of each coefficient function, and combine it with the corresponding dx_i in a particular way:

$$\begin{aligned} d(X^\flat) &= d(a_1 dx_1 + a_2 dx_2 + a_3 dx_3) \\ &= (da_1) \wedge dx_1 + (da_2) \wedge dx_2 + (da_3) \wedge dx_3 \\ &= \left(\frac{\partial a_1}{\partial x_1} dx_1 + \frac{\partial a_1}{\partial x_2} dx_2 + \frac{\partial a_1}{\partial x_3} dx_3 \right) \wedge dx_1 + \left(\frac{\partial a_2}{\partial x_1} dx_1 + \frac{\partial a_2}{\partial x_2} dx_2 + \frac{\partial a_2}{\partial x_3} dx_3 \right) \wedge dx_2 \\ &\quad + \left(\frac{\partial a_3}{\partial x_1} dx_1 + \frac{\partial a_3}{\partial x_2} dx_2 + \frac{\partial a_3}{\partial x_3} dx_3 \right) \wedge dx_3 \\ &= 0 + \frac{\partial a_1}{\partial x_2} dx_2 \wedge dx_1 + \frac{\partial a_1}{\partial x_3} dx_3 \wedge dx_1 + \frac{\partial a_2}{\partial x_1} dx_1 \wedge dx_2 + 0 + \frac{\partial a_2}{\partial x_3} dx_3 \wedge dx_2 \\ &\quad + \frac{\partial a_3}{\partial x_1} dx_1 \wedge dx_3 + \frac{\partial a_3}{\partial x_2} dx_2 \wedge dx_3 + 0 \\ &= \left(\frac{\partial a_3}{\partial x_2} - \frac{\partial a_2}{\partial x_3} \right) dx_2 \wedge dx_3 + \left(\frac{\partial a_1}{\partial x_3} - \frac{\partial a_3}{\partial x_1} \right) dx_3 \wedge dx_1 + \left(\frac{\partial a_2}{\partial x_1} - \frac{\partial a_1}{\partial x_2} \right) dx_1 \wedge dx_2 \end{aligned}$$

where the rule is that $dx_i \wedge dx_j = -dx_j \wedge dx_i$, which is just the alternating condition on forms. So, if you buy the above manipulations on some level, there is a fairly straightforward generalization of the differential which corresponds exactly to the curl in this setting.

⁶which turns out to be necessary to define the gradient anyway

Finally, if $X \in \mathfrak{X}(M)$ is written in local coordinates as $X = a_1 \frac{\partial}{\partial x_1} + a_2 \frac{\partial}{\partial x_2} + a_3 \frac{\partial}{\partial x_3}$, then the divergence is

$$\nabla \cdot X = \frac{\partial a_1}{\partial x_1} + \frac{\partial a_2}{\partial x_2} + \frac{\partial a_3}{\partial x_3},$$

which is a function. But the isomorphism $C^\infty(M) \cong \Omega^3(M)$ described in [Example 2.1.11](#) says that this function corresponds to some 3-form

$$\left(\frac{\partial a_1}{\partial x_1} + \frac{\partial a_2}{\partial x_2} + \frac{\partial a_3}{\partial x_3} \right) dVol_M = \left(\frac{\partial a_1}{\partial x_1} + \frac{\partial a_2}{\partial x_2} + \frac{\partial a_3}{\partial x_3} \right) dx_1 \wedge dx_2 \wedge dx_3$$

because for an appropriate choice of coordinates I can write $dVol_M = dx_1 \wedge dx_2 \wedge dx_3$.

Can we get to this 3-form by applying our generalized differential to the 2-form $\star X^\flat = a_1 dx_2 \wedge dx_3 + a_2 dx_3 \wedge dx_1 + a_3 dx_1 \wedge dx_2$? Yes!

$$\begin{aligned} d(\star X^\flat) &= d(a_1 dx_2 \wedge dx_3 + a_2 dx_3 \wedge dx_1 + a_3 dx_1 \wedge dx_2) \\ &= (da_1) \wedge dx_2 \wedge dx_3 + (da_2) \wedge dx_3 \wedge dx_1 + (da_3) \wedge dx_1 \wedge dx_2 \\ &= \left(\frac{\partial a_1}{\partial x_1} dx_1 + \frac{\partial a_1}{\partial x_2} dx_2 + \frac{\partial a_1}{\partial x_3} dx_3 \right) \wedge dx_2 \wedge dx_3 + \left(\frac{\partial a_2}{\partial x_1} dx_1 + \frac{\partial a_2}{\partial x_2} dx_2 + \frac{\partial a_2}{\partial x_3} dx_3 \right) \wedge dx_3 \wedge dx_1 \\ &\quad + \left(\frac{\partial a_3}{\partial x_1} dx_1 + \frac{\partial a_3}{\partial x_2} dx_2 + \frac{\partial a_3}{\partial x_3} dx_3 \right) \wedge dx_1 \wedge dx_2 \\ &= \frac{\partial a_1}{\partial x_1} dx_1 \wedge dx_2 \wedge dx_3 + 0 + 0 + \frac{\partial a_2}{\partial x_2} dx_2 \wedge dx_3 \wedge dx_1 + 0 + 0 + \frac{\partial a_3}{\partial x_3} dx_3 \wedge dx_1 \wedge dx_2 \\ &= \frac{\partial a_1}{\partial x_1} dx_1 \wedge dx_2 \wedge dx_3 + \frac{\partial a_2}{\partial x_2} dx_1 \wedge dx_2 \wedge dx_3 + \frac{\partial a_3}{\partial x_3} dx_1 \wedge dx_2 \wedge dx_3 \\ &= \left(\frac{\partial a_1}{\partial x_1} + \frac{\partial a_2}{\partial x_2} + \frac{\partial a_3}{\partial x_3} \right) dx_1 \wedge dx_2 \wedge dx_3 \end{aligned}$$

The upshot is that we've now filled in the diagram:

$$\begin{array}{ccccccc} \Omega^0(M) & \xrightarrow{d} & \Omega^1(M) & \xrightarrow{d} & \Omega^2(M) & \xrightarrow{d} & \Omega^3(M) \\ \parallel & & \cong \uparrow \flat & & \cong \uparrow \star^\flat & & \cong \uparrow \cdot dVol_M \\ C^\infty(M) & \xrightarrow{\nabla} & \mathfrak{X}(M) & \xrightarrow{\nabla \times} & \mathfrak{X}(M) & \xrightarrow{\nabla \cdot} & C^\infty(M) \end{array}$$

Of course, I haven't really given a rigorous definition of d (which is called the *exterior derivative*), nor of \wedge (the *wedge product*), but, as we will see, there is a natural, coordinate-free way of defining these things for arbitrary differential forms on arbitrary manifolds which specializes to the calculations above in local coordinates (and hence corresponds to gradient and divergence on manifolds of arbitrary dimension).

Hopefully the fact that div, grad, and curl are all essentially the same operator in this framework helps convince you that differential forms are useful and important. In fact, it gets even better: in this language, the Fundamental Theorem of Calculus, Green's Theorem, and the Divergence Theorem all turn out to be special cases of the same theorem: *Stokes' Theorem for differential forms*.

More than just generalizing essentially all of vector calculus, differential forms also encode topological information (in the form of cohomology). To see this, look again at the diagram above:

- (i) We know from vector calculus that $\nabla \times (\nabla f) = 0$ for any $f \in C^\infty(M)$ and $\nabla \cdot (\nabla \times X) = 0$ for any $X \in \mathfrak{X}(M)$. Since the diagram commutes, this implies that $d \circ d = 0$ regardless of whether you start in $\Omega^0(M)$ or $\Omega^1(M)$ (and in fact this is still true if you start in $\Omega^2(M)$ or $\Omega^3(M)$ since $\Omega^4(M) = \{0\}$). In other words,

$$\Omega^0(M) \xrightarrow{d} \Omega^1(M) \xrightarrow{d} \Omega^2(M) \xrightarrow{d} \Omega^3(M)$$

is a (co)chain complex. More generally, if we add indices to the exterior derivatives to indicate where they start and end,

$$\Omega^0(M) \xrightarrow{d_0} \Omega^1(M) \xrightarrow{d_1} \dots \xrightarrow{d_{n-2}} \Omega^{n-1}(M) \xrightarrow{d_{n-1}} \Omega^n(M)$$

is always a cochain complex on any n -manifold (i.e., $d_k \circ d_{k-1} = 0$ for all k). The usual thing you do when you have a (co)chain complex is to take the (co)homology, and that's also useful in this case. Doing so yields the *de Rham cohomology groups*

$$H_{\text{dR}}^k(M) := \frac{\ker d_k}{\text{im } d_{k-1}}.$$

In fact, as usual with cohomology, the cohomology groups fit together into a graded ring, and the product operation corresponds to the wedge product \wedge . And de Rham cohomology will turn out to be isomorphic to singular (or simplicial) cohomology with real coefficients.

- (ii) In our 3-manifold example, we saw that $\Omega^0(M) = C^\infty(M) \cong \Omega^3(M)$ and $\Omega^1(M) \cong \mathfrak{X}(M) \cong \Omega^2(M)$. In general, if M is an n -dimensional manifold, then it will turn out that $\Omega^k(M) \cong \Omega^{n-k}(M)$ for all $k = 0, \dots, n$. Even better, this isomorphism descends to the de Rham cohomology groups and we will have that

$$H_{\text{dR}}^k(M) \cong H_{\text{dR}}^{n-k}(M)$$

for all k , which is the appropriate version of *Poincaré duality* for de Rham cohomology.

2.3 Tensor Algebras and Tensor Fields

In order to give a more rigorous definition of differential forms (and in particular to get them to form a graded algebra), as well as more general tensor fields, we need to do some (multilinear) algebra and talk about tensor algebras and exterior algebras.

Before diving in, let me just give you my perspective on tensors. Basically, the point is that tensors are the right tool for turning multilinear algebra into linear algebra. More precisely, suppose we have three vector spaces U , V , and W , and a map $F : U \times V \rightarrow W$ which is multilinear (or, really, bilinear in this case). Again, this just means that F is linear in each factor:

$$F(au_1 + bu_2, v) = aF(u_1, v) + bF(u_2, v) \quad \text{and} \quad F(u, cv_1 + dv_2) = cF(u, v_1) + dF(u, v_2).$$

Example 2.3.1. Any choice of inner product on \mathbb{R}^n defines a bilinear map $\mathbb{R}^n \times \mathbb{R}^n \rightarrow \mathbb{R}$ by $(u, v) \mapsto \langle u, v \rangle$. Here $U = V = \mathbb{R}^n$ and $W = \mathbb{R}$.

Example 2.3.2. Let $U = \mathbb{R}^n$, $V = (\mathbb{R}^n)^{n-1}$ and define a map $\mathbb{R}^n \times (\mathbb{R}^n)^{n-1} \cong (\mathbb{R}^n)^n \rightarrow \mathbb{R}$ by

$$(u, (v_1, \dots, v_{n-1})) \mapsto \det [u \quad v_1 \quad \dots \quad v_{n-1}].$$

This is also bilinear.

Example 2.3.3. Let $U = V = W = \mathbb{R}^3$ and define the map $\mathbb{R}^3 \times \mathbb{R}^3 \rightarrow \mathbb{R}^3$ by $(u, v) \mapsto u \times v$, which is again bilinear.

Returning to the general setting, we have vector space U, V , and W and a multilinear map $F : U \times V \rightarrow W$. Now suppose $w_0 \in W$ and we want to solve a problem of the form $F(u, v) = w_0$. Since F is multilinear, you might hope that this is somehow just a linear algebra problem, which would be solvable, at least in principle. But, at least as stated, this is very much *not* a linear problem. . .

Example 2.3.4. Continuing with Example 2.3.3, where $F : \mathbb{R}^3 \times \mathbb{R}^3 \rightarrow \mathbb{R}^3$ is given by $F(u, v) = u \times v$, suppose we want to solve $F(u, v) = (x_0, y_0, z_0)$. Since $\mathbb{R}^3 \times \mathbb{R}^3 \cong \mathbb{R}^6$, we can think of F as a map $\mathbb{R}^6 \rightarrow \mathbb{R}^3$ given by

$$(u_1, u_2, u_3, v_1, v_2, v_3) \mapsto u \times v = (u_2 v_3 - u_3 v_2, u_3 v_1 - u_1 v_3, u_1 v_2 - u_2 v_1).$$

So our problem is to solve

$$(u_2 v_3 - u_3 v_2, u_3 v_1 - u_1 v_3, u_1 v_2 - u_2 v_1) = (x_0, y_0, z_0)$$

for $(u_1, u_2, u_3, v_1, v_2, v_3)$, which is obviously a *quadratic* system of equations, not a linear system.

To reiterate, for me⁷ the point of tensor products is to turn multilinear maps and problems into linear maps and problems.

Definition 2.3.5. Given vector spaces U and V , define the *tensor product* $U \otimes V$ to be the vector space whose elements are linear combinations of terms of the form $u \otimes v$ for $u \in U$ and $v \in V$ so that

- (i) $(u_1 + u_2) \otimes v = u_1 \otimes v + u_2 \otimes v$
- (ii) $u \otimes (v_1 + v_2) = u \otimes v_1 + u \otimes v_2$
- (iii) $a(u \otimes v) = (au) \otimes v = u \otimes (av)$

for any $u, u_1, u_2 \in U$, $v, v_1, v_2 \in V$ and $a \in \mathbb{R}$.

Example 2.3.6. If $U = \mathbb{R}^m$ and $V = \mathbb{R}^n$, then we can represent $u \otimes v$ by the $m \times n$ matrix

$$uv^T = \begin{bmatrix} u_1 \\ \vdots \\ u_m \end{bmatrix} \begin{bmatrix} v_1 & \dots & v_n \end{bmatrix} = [u_i v_j]_{i,j}.$$

Here's the key theorem (or, if you start from a different perspective, the below theorem is the definition):

Theorem 2.3.7 (Universal Property of the Tensor Product). *If $\phi : U \times V \rightarrow U \otimes V$ is the map given by $(u, v) \mapsto u \otimes v$ and $F : U \times V \rightarrow W$ is bilinear, then there exists a unique linear map $\tilde{F} : U \otimes V \rightarrow W$ making the following diagram commute:*

$$\begin{array}{ccc} U \times V & \xrightarrow{F} & W \\ \phi \downarrow & \nearrow \tilde{F} & \\ U \otimes V & & \end{array}$$

Moreover, this uniquely characterizes $U \otimes V$: any vector space satisfying this property must be isomorphic to $U \otimes V$.

⁷I want to emphasize here that this is just my perspective: I'm not claiming that everybody would agree with this characterization.

Example 2.3.8. Continuing Examples 2.3.3 and 2.3.4, recall that $F: \mathbb{R}^3 \times \mathbb{R}^3 \rightarrow \mathbb{R}^3$ is given by $F(u, v) = u \times v$, and Theorem 2.3.7 tells us there must be a linear map $\tilde{F}: \mathbb{R}^3 \otimes \mathbb{R}^3 \rightarrow \mathbb{R}^3$ so that

$$(\tilde{F} \circ \phi)(u, v) = F(u, v) = u \times v.$$

Representing $u \otimes v$ by uv^T as in Example 2.3.6, we see that $u \otimes v$ corresponds to the 3×3 matrix

$$\begin{bmatrix} u_1v_1 & u_1v_2 & u_1v_3 \\ u_2v_1 & u_2v_2 & u_2v_3 \\ u_3v_1 & u_3v_2 & u_3v_3 \end{bmatrix}.$$

In general, $\mathbb{R}^3 \otimes \mathbb{R}^3 \cong \text{Mat}_{3 \times 3}(\mathbb{R}) \cong \mathbb{R}^9$, so we can flatten this matrix to get a representation by a 9-dimensional vector, namely

$$\begin{bmatrix} u_1v_1 \\ u_1v_2 \\ u_1v_3 \\ u_2v_1 \\ u_2v_2 \\ u_2v_3 \\ u_3v_1 \\ u_3v_2 \\ u_3v_3 \end{bmatrix}.$$

So then $\tilde{F}: \mathbb{R}^9 \rightarrow \mathbb{R}^3$ is supposed to be linear, and hence must be represented by a 3×9 matrix A so that

$$A \begin{bmatrix} u_1v_1 \\ u_1v_2 \\ u_1v_3 \\ u_2v_1 \\ u_2v_2 \\ u_2v_3 \\ u_3v_1 \\ u_3v_2 \\ u_3v_3 \end{bmatrix} = \begin{bmatrix} u_2v_3 - u_3v_2 \\ u_3v_1 - u_1v_3 \\ u_1v_2 - u_2v_1 \end{bmatrix}.$$

Written out in this excruciating detail, it's now pretty obvious what A has to be:

$$A = \begin{bmatrix} 0 & 0 & 0 & 0 & 0 & 1 & 0 & -1 & 0 \\ 0 & 0 & -1 & 0 & 0 & 0 & 1 & 0 & 0 \\ 0 & 1 & 0 & -1 & 0 & 0 & 0 & 0 & 0 \end{bmatrix}.$$

Proposition 2.3.9. If u_1, \dots, u_m is a basis for U and v_1, \dots, v_n is a basis for V , then $\{e_i \otimes f_j\}_{i,j}$ is a basis for $U \otimes V$. In particular, $\dim U \otimes V = mn$.

Clearly, we've paid a price for turning bilinear maps into linear maps: whereas $\dim(U \times V) = m + n$, we have $\dim(U \otimes V) = mn$, which is usually a lot bigger. So we turn multilinear algebra into linear algebra at the cost of having to work in much higher dimensions.

Notation. We will use $V^{\otimes k}$ to denote the k -fold tensor product of a vector space V with itself; that is

$$\underbrace{V \otimes \cdots \otimes V}_k = V \otimes (V \otimes (\cdots \otimes (V \otimes V)))$$

Definition 2.3.10. The *tensor space of type* (r, s) associated with a vector space V is the vector space

$$V_r^s := V^{\otimes r} \otimes (V^*)^{\otimes s}.$$

Elements of V_r^s are called (r, s) -*tensors*.

Example 2.3.11. An inner product $\langle \cdot, \cdot \rangle$ on V is a symmetric $(0, 2)$ -tensor on V . Why? Well, $\langle \cdot, \cdot \rangle$ defines a bilinear map $V \times V \rightarrow \mathbb{R}$, and hence by [Theorem 2.3.7](#) a (unique) linear map $V \otimes V \rightarrow \mathbb{R}$. In other words, an inner product determines an element of $(V \otimes V)^* \cong V^* \otimes V^* = V_0^2$.

Exercise 2.3.12. Prove that $(V \otimes V)^* \cong V^* \otimes V^*$

Example 2.3.13. A linear transformation $F: V \rightarrow V$ (or, if you like, a square matrix) can be interpreted as a $(1, 1)$ -tensor. Why? F induces a bilinear map $V^* \times V \rightarrow \mathbb{R}$ as follows: for $(\rho, v) \in V^* \times V$,

$$(\rho, v) \mapsto \rho(F(v)).$$

Again, [Theorem 2.3.7](#) implies that this bilinear map corresponds to a linear map $V^* \otimes V \rightarrow \mathbb{R}$; that is, an element of the dual space $(V^* \otimes V)^* \cong V \otimes V^* = V_1^1$.

Definition 2.3.14. The direct sum $\mathcal{T}(V) := \bigoplus_{r,s \geq 0} V_r^s$, where V_0^0 is the ground field of V , is called the *tensor algebra* of V .

Remark 2.3.15. Terminology varies. Often *tensor algebra* only refers to the algebra of $(\cdot, 0)$ -tensors, namely $\bigoplus_{r \geq 0} V_r^0 = \bigoplus_{r \geq 0} V^{\otimes r}$.

In general, $\mathcal{T}(V)$ is noncommutative, associative, and (bi-)graded.

Definition 2.3.16. The *tensor bundle of type* (r, s) over a manifold M is

$$\mathcal{T}_r^s(M) := \bigsqcup_{p \in M} (T_p M)_r^s,$$

which has a projection map $\pi: \mathcal{T}_r^s(M) \rightarrow M$ sending a tensor based at a point to the point.

Exercise 2.3.17. If M is an n -dimensional manifold, show that $\mathcal{T}_r^s(M)$ is a smooth manifold of dimension $n + n^{r+s}$.

A *tensor field* of type (r, s) on M is a smooth section of the tensor bundle $\mathcal{T}_r^s(M)$; i.e., a smooth map $\tau: M \rightarrow \mathcal{T}_r^s(M)$ so that $\tau(p) \in (T_p M)_r^s$.

Example 2.3.18. A vector field on M is (by definition) a $(1, 0)$ -tensor field on M .

Example 2.3.19. A 1-form on M is (again, by definition) a $(0, 1)$ -tensor field.

Example 2.3.20. A Riemannian metric (which we've informally defined to be a smooth choice of inner product on each tangent space) is a (symmetric) $(0, 2)$ -tensor field on M . More precisely:

Definition 2.3.21. A *Riemannian metric* on a manifold M is a smooth $(0, 2)$ -tensor field g on M so that, for each $p \in M$, $g_p = g(p)$ satisfies the axioms of an inner product on $T_p(M)$; that is, in addition to being bilinear (which is guaranteed by the fact it is a $(0, 2)$ -tensor), it must be symmetric ($g_p(u, v) = g_p(v, u)$) and positive-definite ($g_p(v, v) \geq 0$ with equality if and only if $v = 0$).

Note: when I write something like $g_p(u, v)$ I'm technically thinking of g_p as the associated bilinear map $T_p M \times T_p M \rightarrow \mathbb{R}$, but in coordinates we also often use notation like $g = \sum_{i,j} g_{ij} dx_i \otimes dx_j$, which more closely matches up with the idea of g being a $(0, 2)$ -tensor field.

Example 2.3.22. Consider the Riemannian metric $g = \frac{1}{y^2}(dx \otimes dx + dy \otimes dy)$ (also commonly written as $ds^2 = \frac{1}{y^2}(dx^2 + dy^2)$) on $H = \{(x, y) \in \mathbb{R}^2 : y > 0\}$.⁸ This is the Poincaré half-plane model of the hyperbolic plane. In general, we can compute the length of a curve $\gamma(t) = (\gamma_1(t), \gamma_2(t))$ by integrating the length of $\gamma'(t)$ along the curve. Since $\gamma'(t) \in T_{\gamma(t)}H$, its length is

$$\sqrt{g_{\gamma(t)}(\gamma'(t), \gamma'(t))} = \sqrt{\frac{\gamma'(t) \cdot \gamma'(t)}{\gamma_2(t)^2}} = \frac{\|\gamma'(t)\|_E}{\gamma_2(t)},$$

where \cdot is the standard dot product on the plane and $\|\cdot\|_E$ is the Euclidean norm (I can just use $\gamma_2(t)$ in the denominator, rather than $|\gamma_2(t)|$, because the second coordinate is always positive in H).

It will turn out that the geodesics in this metric are the vertical lines and semicircles centered at points on the x -axis; see [Figure 2.1](#). Let's compute distances between points on these geodesics.

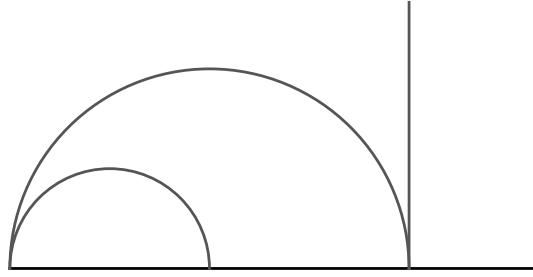


Figure 2.1: Some geodesics in the Poincaré upper half-plane model of the hyperbolic plane.

For the vertical lines, we can parametrize the line through $(x_0, 0)$ by $\gamma(t) = (x_0, t)$. So then $\gamma'(t) = (0, 1)$ and the distance between (x_0, y_0) and (x_0, y_1) is

$$\int_{y_0}^{y_1} \sqrt{g_{\gamma(t)}(\gamma'(t), \gamma'(t))} dt = \int_{y_0}^{y_1} \frac{1}{t} dt = \ln y_1 - \ln y_0 = \ln \left(\frac{y_1}{y_0} \right).$$

Notice, in particular, that the x -axis is infinitely far away from any point in H .

On the other hand, the semicircle of radius r centered at $(x_0, 0)$ can be parametrized by $\gamma(t) = (x_0 + r \cos t, r \sin t)$. Now $\gamma'(t) = r(-\sin t, \cos t)$, so the distance between $\gamma(\theta_0)$ and $\gamma(\theta_1)$ is

$$\int_{\theta_0}^{\theta_1} \sqrt{g_{\gamma(t)}(\gamma'(t), \gamma'(t))} dt = \int_{\theta_0}^{\theta_1} \frac{r}{r \sin t} dt = \int_{\theta_0}^{\theta_1} \csc t dt = \operatorname{arctanh}(\cos \theta_0) - \operatorname{arctanh}(\cos \theta_1).$$

⁸In undergraduate differential geometry courses like MATH 474, the Riemannian metric is more often called the First Fundamental Form.

2.4 Exterior Algebras and Differential Forms

We just saw that 1-forms are $(0, 1)$ -tensor fields, but in general it's not quite natural to think of k -forms as tensor fields because we need to incorporate the alternating condition. We could do this by considering the subset of alternating tensor fields, but it turns out to be more natural to take a quotient of the tensor algebra to get something called the *exterior algebra*.

Specifically, just as the tensor product is a way of turning multilinear maps into linear maps, the *exterior product* will turn out to be the way to turn alternating multilinear maps into linear maps. Even though we haven't actually defined the exterior product yet, it might seem silly to introduce another new product; after all, alternating multilinear maps are multilinear so the tensor product *already* turns them into linear maps.

That being the case, the exterior product does this in much lower dimension, so it's still worthwhile. At the most extreme level, you can see this because the natural space in which the tensor product lives (as a binary operation) is the tensor algebra $\mathcal{T}(V)$ (or, if you like, only the $(r, 0)$ part $V_0^0 \otimes V_1^0 \otimes V_2^0 \otimes \dots$), which is always infinite-dimensional. Whereas the exterior product will be a binary operation on the exterior algebra, which will turn out to be finite-dimensional if V is. And, if you weren't already aware, finite-dimensional vector spaces are *much, much nicer* than infinite-dimensional vector spaces: it's always worthwhile to make the effort to work in finite dimensions if at all possible.

Definition 2.4.1. Let $C(V) = \bigoplus_{k \geq 0} V_k^0$ be the algebra of $(\cdot, 0)$ tensors on a vector space V and let $\mathcal{I}(V)$ be the two-sided ideal generated by elements of the form $v \otimes v$ for $v \in V$. Then the *exterior algebra* of V is the quotient

$$\wedge(V) := C(V)/\mathcal{I}(V).$$

This is a graded algebra with product denoted by \wedge , which is just the product induced by \otimes : the residue class of $v_1 \otimes v_2 \otimes \dots \otimes v_k$ is denoted $v_1 \wedge v_2 \wedge \dots \wedge v_k$.

The grading is given by the degree of the wedge product, since

$$\wedge(V) = \bigoplus_{k \geq 0} \wedge^k(V),$$

where $\wedge^k(V) = V_k^0/\mathcal{I}_k$, and $\mathcal{I}_k = \mathcal{I}(V) \cap V_k^0$.

Proposition 2.4.2. If $a \in \wedge^k(V)$ and $b \in \wedge^\ell(V)$, then $b \wedge a = (-1)^{k\ell} a \wedge b$.

Exercise 2.4.3. Prove Proposition 2.4.2.

Corollary 2.4.4. $v_{\sigma(1)} \wedge \dots \wedge v_{\sigma(k)} = \text{sgn}(\sigma)v_1 \wedge \dots \wedge v_k$ for any permutation σ of $\{1, \dots, k\}$.

In particular, this says that we can always rearrange any wedge product to have the terms in any order we like. For example, if the terms are indexed (e.g., if they come from some ordered basis), we can always make the indices appear in increasing order at the cost of a sign.

Proposition 2.4.5. If $\{e_1, \dots, e_n\}$ is a basis for V , then

$$\{e_{i_1} \wedge \dots \wedge e_{i_k} : 1 \leq i_1 < i_2 < \dots < i_k \leq n\}$$

is a basis for $\wedge^k(V)$. In particular,

$$\dim \wedge^k(V) = \begin{cases} \binom{n}{k} & \text{if } 0 \leq k \leq n \\ 0 & \text{else.} \end{cases}$$

Hence,

$$\dim \wedge(V) = \sum_{k \geq 0} \dim \wedge^k(V) = \sum_{k=0}^n \binom{n}{k} = 2^n.$$

Exercise 2.4.6. Prove Proposition 2.4.5.

This is exactly the finite-dimensionality described above. Even if we're just interested in alternating, bilinear maps $V \times V \rightarrow W$, $\dim \wedge^2(V) = \binom{n}{2}$ is less than half of $\dim V^{\otimes 2} = n^2$, so working with the exterior power rather than the tensor power means we're doing linear algebra in half the dimensions, which is almost always worthwhile computationally.

As you would expect, the exterior products satisfy a universal property analogous to the one satisfied by tensor products:

Theorem 2.4.7 (Universal Property of the Exterior Product). *If $\phi: V^k \rightarrow \wedge^k(V)$ is the map given by $(v_1, \dots, v_k) \mapsto v_1 \wedge \dots \wedge v_k$ and $F: V^k \rightarrow W$ is an alternating multilinear map, then there exists a unique linear map $\tilde{F}: \wedge^k(V) \rightarrow W$ making the following diagram commute:*

$$\begin{array}{ccc} \underbrace{V \times \dots \times V}_k & \xrightarrow{F} & W \\ \downarrow \phi & \nearrow \tilde{F} & \\ \wedge^k(V) & & \end{array}$$

Again, this uniquely characterizes $\wedge^k(V)$ up to isomorphism.

Example 2.4.8. Suppose $V = \mathbb{R}^n$, and suppose $F: \mathbb{R}^n \rightarrow \mathbb{R}$ is alternating and multilinear. We know that this factors through a linear map $\wedge^n(\mathbb{R}^n) \rightarrow \mathbb{R}$; since $\dim \wedge^n \mathbb{R}^n = \binom{n}{n} = 1$, we know that $\wedge^n \mathbb{R}^n \cong \mathbb{R}$, and so (once we've chosen a basis for $\wedge^n \mathbb{R}^n$, which witnesses the isomorphism $\wedge^n \mathbb{R}^n \cong \mathbb{R}$), \tilde{F} has to be of the form $x \mapsto ax$ for some scalar a . We can already see that this is going to imply that F is unique up to the choice of the scalar a .

Let's work out what $\phi: (\mathbb{R}^n)^n \rightarrow \wedge^n \mathbb{R}^n$ is. Let e_1, \dots, e_n be the standard basis for \mathbb{R}^n , so that each $v_i = \sum_{j_1=1}^n a_{ij_1} e_{j_1}$. Then repeatedly applying Corollary 2.4.4 and the alternating property of the wedge

product yields

$$\begin{aligned}
\phi(v_1, \dots, v_n) &= \phi\left(\sum_{j_1=1}^n a_{1j_1} e_{j_1}, \dots, \sum_{j_n=1}^n a_{nj_n} e_{j_n}\right) \\
&= \left(\sum_{j_1=1}^n a_{1j_1} e_{j_1}\right) \wedge \cdots \wedge \left(\sum_{j_n=1}^n a_{nj_n} e_{j_n}\right) \\
&= \sum_{j_1=1}^n \cdots \sum_{j_n=1}^n a_{1j_1} \cdots a_{nj_n} e_{j_1} \wedge \cdots \wedge e_{j_n} \\
&= \sum_{\sigma \in S_n} \left[\left(\prod_{i=1}^n a_{i\sigma(i)} \right) e_{\sigma(1)} \wedge \cdots \wedge e_{\sigma(n)} \right] \\
&= \sum_{\sigma \in S_n} \left[\operatorname{sgn}(\sigma) \left(\prod_{i=1}^n a_{i\sigma(i)} \right) e_1 \wedge \cdots \wedge e_n \right] \\
&= \left[\sum_{\sigma \in S_n} \operatorname{sgn}(\sigma) \left(\prod_{i=1}^n a_{i\sigma(i)} \right) \right] e_1 \wedge \cdots \wedge e_n \\
&= \det [v_1 \ \cdots \ v_n] e_1 \wedge \cdots \wedge e_n,
\end{aligned}$$

where we just recognized the Leibniz formula for the determinant in the last equality.

Since $\{e_1 \wedge \cdots \wedge e_n\}$ is our preferred basis for $\wedge^n \mathbb{R}^n$, inducing the isomorphism $\mathbb{R} \xrightarrow{\cong} \wedge^n \mathbb{R}^n$ given by $1 \mapsto e_1 \wedge \cdots \wedge e_n$, we see that

$$F(v_1, \dots, v_n) = (\tilde{F} \circ \phi)(v_1, \dots, v_n) = \tilde{F}(\phi(v_1, \dots, v_n)) = \tilde{F}(\det [v_1 \ \cdots \ v_n] e_1 \wedge \cdots \wedge e_n) = a \det [v_1 \ \cdots \ v_n]$$

for some $a \in \mathbb{R}$, which (re)proves [Theorem 2.1.3](#), which says that the determinant is, up to scale, the unique alternating multilinear map $(\mathbb{R}^n)^n \rightarrow \mathbb{R}$.

If we let $W = \mathbb{R}$ (or whatever the base field is) in [Theorem 2.4.7](#), then we see that alternating multilinear functionals on V^k correspond precisely to linear functionals on $\wedge^k V$; that is, to elements of the dual space $(\wedge^k V)^*$. But of course this is essentially how we “defined” differential k -forms in [Definition 2.1.4](#): at each point a k -form should exactly be an alternating multilinear functional on the product of k copies of the tangent space at that point.

Definition 2.4.9. The *exterior k -bundle* on a manifold M is

$$\wedge^k(M) := \bigsqcup_{p \in M} \wedge^k((T_p M)^*)$$

and the *exterior algebra bundle* over M is

$$\wedge(M) := \bigsqcup_{p \in M} \wedge((T_p M)^*) .$$

Both of these are smooth manifolds.

A *differential k-form* on M is a smooth section of $\bigwedge_n^k(M)$; that is, a smooth map $\omega: M \rightarrow \bigwedge_n^k(M)$ so that $\omega(p) \in \bigwedge^k((T_p M)^*)$. The space of k -forms on M is denoted $\Omega^k(M)$, and we let $\Omega^*(M) := \bigoplus_{k=0}^n \Omega^k(M)$, which is the algebra of differential forms on M .

If you've been reading very closely, you might have noticed that I played fast and loose with parentheses and stars above. The discussion in the paragraph before [Definition 2.4.9](#) suggests that, if ω is a k -form (that is, an alternating, multilinear map $\mathfrak{X}(M)^k \rightarrow C^\infty(M)$ according to our informal definition), then at a point $p \in M$ we should expect that $\omega(p) \in (\bigwedge^k(T_p M))^*$. But I've just defined $\omega(p)$ as an element of $\bigwedge^k((T_p M)^*)$! So what's going on?

First of all, $\bigwedge^k((T_p M)^*)$ is (at least to me) conceptually nicer than $(\bigwedge^k(T_p M))^*$: it's an exterior power itself, rather than being the dual of an exterior power, and so it obviously fits into an exterior algebra, namely $\bigwedge((T_p M)^*)$. So that's the reason for wanting to define exterior bundles in terms of $\bigwedge^k((T_p M)^*)$ rather than $(\bigwedge^k(T_p M))^*$.

Of course, the reason I can do this is because they're isomorphic:

Lemma 2.4.10. *Suppose V is a finite-dimensional vector space. Then*

$$\bigwedge^k(V^*) \cong (\bigwedge^k V)^*,$$

and therefore

$$\bigwedge(V^*) \cong (\bigwedge(V))^*.$$

The strategy here is to find a nondegenerate pairing between $\bigwedge(V^*)$ and $\bigwedge(V)$. Recall that a *nondegenerate pairing* of vector space V and W is a bilinear map

$$(\cdot, \cdot): V \times W \rightarrow \mathbb{R}$$

with the property that, for any nonzero $w \in W$, there exists $v \in V$ so that $(v, w) \neq 0$ (and similarly when you swap the roles of v and w). Notice, in particular, that a pairing corresponds uniquely to a linear map $V \otimes W \rightarrow \mathbb{R}$; that is, an element of $(V \otimes W)^*$.

If there is a nondegenerate pairing of V and W , then it induces injective maps $\phi: V \rightarrow W^*$ and $\psi: W \rightarrow V^*$ given by

$$\phi(v)(w) := (v, w) =: \psi(w)(v).$$

Example 2.4.11. An inner product $\langle \cdot, \cdot \rangle$ on V is a nondegenerate pairing of V with itself: after all, if $v \neq 0$, then $\langle v, v \rangle \neq 0$ by the positivity of the inner product. Hence, the inner product induces an injective map $V \rightarrow V^*$ which turns out to be an isomorphism if V is finite-dimensional (or more generally if we take V^* to be the continuous dual space rather than the algebraic dual space); this isomorphism is the content of the Riesz representation theorem.

In finite dimensions, we know that $V \cong V^*$ and $W \cong W^*$, so the composition $V \xrightarrow{\phi} W^* \cong W \xrightarrow{\psi} V^*$ gives an injective (and therefore also surjective) map $V \rightarrow V^*$, which implies that ϕ and ψ had to be bijective, and therefore isomorphisms. In particular, this implies that $V \cong W^*$ (and therefore also $W \cong V^*$ and indeed $V \cong W$).

So the strategy for proving [Lemma 2.4.10](#) is to find a nondegenerate pairing of $\bigwedge(V^*)$ and $\bigwedge(V)$, which will imply the result.⁹

⁹At a more basic level, it's not hard to see that these vector spaces have the same dimension, and hence must be abstractly isomorphic, but it's helpful to make this isomorphism more concrete with the pairing.

Proof of Lemma 2.4.10. To find the pairing $(\cdot, \cdot) : \wedge^k(V^*) \times \wedge^k(V) \rightarrow \mathbb{R}$, suppose first that $\alpha \in \wedge^k(V^*)$ and $b \in \wedge^k(V)$ are *decomposable*, meaning that $\alpha = \alpha_1 \wedge \cdots \wedge \alpha_k$ for $\alpha_1, \dots, \alpha_k \in V^*$ and $b = b_1 \wedge \cdots \wedge b_k$ for $b_1, \dots, b_k \in V$. Then define

$$(\alpha, b) := \det \left([\alpha_i(b_j)]_{i,j} \right). \quad (2.2)$$

Since arbitrary elements of $\wedge^k(V^*)$ and $\wedge^k(V)$ are linear combinations of decomposable elements, we define $(\cdot, \cdot) : \wedge^k(V^*) \times \wedge^k(V) \rightarrow \mathbb{R}$ by extending the above linearly.

It is straightforward (but somewhat tedious) to check that (\cdot, \cdot) is nondegenerate, and thus induces the desired isomorphism $\wedge^k(V^*) \xrightarrow{\cong} (\wedge^k(V))^*$. Doing this for each k shows that the exterior algebra

$$\wedge(V^*) = \bigoplus_{k=0}^{\dim V} \wedge^k(V^*) \cong \bigoplus_{k=0}^{\dim V} (\wedge^k(V))^* \cong \left(\bigoplus_{k=0}^{\dim V} \wedge^k(V) \right)^* = (\wedge(V))^*.$$

□

Notice that this argument—and especially (2.2)—really validates the idea that all alternating multilinear maps are, in some sense, determinants.

2.4.1 Differential Forms in Local Coordinates

Definition 2.4.9 is not the easiest thing to compute with, so let's try to understand what differential forms look like in local coordinates. Suppose M is an n -manifold, $\omega \in \Omega^k(M)$ is a k -form, and $p \in M$. Let (U, ϕ) be a local coordinate chart in a neighborhood of p , and define the functions $x_i : \phi(U) \rightarrow \mathbb{R}$ by $x_i(\phi(a_1, \dots, a_n)) = a_i$. In other words, the function x_i reports the i th local coordinate of a point in this coordinate chart.

Now, each function x_i has a differential dx_i which (as in [Example 2.1.7](#)) is a 1-form on $\phi(U) \subset M$. In particular, at p the differential $(dx_i)_p : T_p M \rightarrow T_{x_i(p)} \mathbb{R} \cong \mathbb{R}$ is a linear functional on $T_p M$; that is, an element of the dual space $(T_p M)^*$.

If $\left\{ \frac{\partial}{\partial x_1}, \dots, \frac{\partial}{\partial x_n} \right\}$ is the local coordinate basis for $T_p M$ associated to our chart, I claim that $\{(dx_1)_p, \dots, (dx_n)_p\}$ gives the corresponding dual basis for $(T_p M)^*$: by [Lemma 1.3.4](#),

$$(dx_i)_p \left(\frac{\partial}{\partial x_j} \right) = \frac{\partial}{\partial x_j}(x_i) = \delta_{ij}.$$

In what follows I'm going to drop the subscript p and just say $\{dx_1, \dots, dx_n\}$ is a basis for $(T_p M)^* = \wedge^1((T_p M)^*)$. More generally, as in [Proposition 2.4.5](#), $\{dx_{i_1} \wedge \cdots \wedge dx_{i_k} : 1 \leq i_1 < \cdots < i_k \leq n\}$ gives a basis for $\wedge^k((T_p M)^*)$, and hence at the point p our k -form ω can be written as

$$\omega_p = \sum_I a_I dx_I,$$

where $I = (i_1, \dots, i_k)$ is any k -tuple of distinct, sorted indices and dx_I is shorthand for $dx_{i_1} \wedge \cdots \wedge dx_{i_k}$.

Example 2.4.12. On \mathbb{R}^2 we have global coordinates, so every $\omega \in \Omega^1(\mathbb{R}^2)$ is of the form $\omega = f dx + g dy$ for smooth functions $f, g \in C^\infty(\mathbb{R}^2)$. So if $\omega = f dx + g dy$ and $\eta = h dx + m dy$ are two 1-forms, we will have

$$\omega \wedge \eta = (f dx + g dy) \wedge (h dx + m dy) = (fm - gh) dx \wedge dy.$$

In particular,

$$(\omega \wedge \eta)(U, V) = (\omega \wedge \eta) \left(u_1 \frac{\partial}{\partial x} + u_2 \frac{\partial}{\partial y}, v_1 \frac{\partial}{\partial x} + v_2 \frac{\partial}{\partial y} \right) = (fm - gh)(u_1 v_2 - u_2 v_1).$$

Example 2.4.13. On \mathbb{R}^3 , if $\omega, \eta \in \Omega^1(\mathbb{R}^3)$ with $\omega = f dx + g dy + h dz$ and $\eta = a dx + b dy + c dz$, then

$$\omega \wedge \eta = (gc - hb)dy \wedge dz + (ha - fc)dz \wedge dx + (fb - ga)dx \wedge dy.$$

Compare with the cross product

$$\omega^\# \times \eta^\# = \left(f \frac{\partial}{\partial x} + g \frac{\partial}{\partial y} + h \frac{\partial}{\partial z} \right) \times \left(a \frac{\partial}{\partial x} + b \frac{\partial}{\partial y} + c \frac{\partial}{\partial z} \right) = (gc - hb) \frac{\partial}{\partial x} + (ha - fc) \frac{\partial}{\partial y} + (fb - ga) \frac{\partial}{\partial z}.$$

Also, if $\mu = p dy \wedge dz + q dz \wedge dx + r dx \wedge dy$, then

$$\omega \wedge \mu = (f dx + g dy + h dz) \wedge (p dy \wedge dz + q dz \wedge dx + r dx \wedge dy) = (fp + gq + hr)dx \wedge dy \wedge dz.$$

Compare with the dot product

$$\omega^\# \cdot (\star \mu)^\# = \left(f \frac{\partial}{\partial x} + g \frac{\partial}{\partial y} + h \frac{\partial}{\partial z} \right) \times \left(p \frac{\partial}{\partial x} + q \frac{\partial}{\partial y} + r \frac{\partial}{\partial z} \right) = fp + gq + hr.$$

Locally, any manifold looks like \mathbb{R}^n , so these computations work just as well in local coordinates on 2- and 3-manifolds.

2.5 The Exterior Derivative

We've already seen some calculations of the exterior derivative in local coordinates in [Section 2.2](#), but to give a coordinate-free definition of the exterior derivative we need to talk about derivations.

Definition 2.5.1. If V is a vector space, an endomorphism $D : \wedge(V) \rightarrow \wedge(V)$ of the exterior algebra is

- a *derivation* if $D(u \wedge v) = (Du) \wedge v + u \wedge (Dv)$
- an *anti-derivation* if $D(u \wedge v) = (Du) \wedge v + (-1)^k u \wedge (Dv)$ for $u \in \wedge^k(V)$ (notice that this implies that $D(v_1 \wedge \cdots \wedge v_m) = \sum_{i=1}^m (-1)^{i-1} v_1 \wedge \cdots \wedge v_{i-1} \wedge (Dv_i) \wedge v_{i+1} \wedge \cdots \wedge v_m$ for any $v_1, \dots, v_m \in V = \wedge^1(V)$)
- of *degree k* if $D : \wedge^i(V) \rightarrow \wedge^{i+k}(V)$ for all i .

(This definition extends to any (graded) algebra by substituting the product in the algebra for \wedge in the above.)

Example 2.5.2. We saw on [page 8](#) that tangent vectors can be viewed as (ungraded) derivations on the algebra \mathcal{D}_p of functions which are smooth in a neighborhood of a point $p \in M$. More generally, a vector field $X \in \mathfrak{X}(M)$ is a derivation on the algebra $C^\infty(M)$ because it satisfies the Leibniz rule $X(fg) = (Xf)g + f(Xg)$.

There's no degree in either of these cases because \mathcal{D}_p and $C^\infty(M)$ are ungraded.

Definition 2.5.3. The *exterior derivative* $d : \Omega^*(M) \rightarrow \Omega^*(M)$ is the unique anti-derivation of degree +1 so that

- (i) For all $f \in C^\infty(M) = \Omega^0(M)$, df is the differential of f .
- (ii) $d \circ d = 0$.

Proposition 2.5.4. *There is such a creature and it is unique.*

The idea is that we kind of know what such an thing should look like in local coordinates around a point $p \in M$: any k -form $\omega \in \Omega^k(M)$ can be written in local coordinates as

$$\omega_p = \sum_I f_I dx_I.$$

So then define

$$(d\omega)_p := \sum_I (df_I) \wedge dx_I. \quad (2.3)$$

Example 2.5.5. Consider $\omega \in \Omega^1(\mathbb{R}^3)$ given by $\omega = z dy + xy dz$. Then

$$\begin{aligned} d\omega &= (dz) \wedge dy + (d(xy)) \wedge dz = dz \wedge dy + \left(\frac{\partial(xy)}{\partial x} dx + \frac{\partial(xy)}{\partial y} dy + \frac{\partial(xy)}{\partial z} dz \right) \wedge dz \\ &= dz \wedge dy + y dx \wedge dz + x dy \wedge dz = y dx \wedge dz + (x - 1) dy \wedge dz. \end{aligned}$$

I claim that (2.3) is the only way to get an anti-derivation of degree +1 that satisfies (i) and (ii) in local coordinates. Why? If d satisfies (i) and (ii), then $dx_i = d(x_i)$ is the result of applying d to the 0-form x_i (by (i)), so $d(dx_i) = (d \circ d)(x_i) = 0$ (by (ii)). But then the anti-derivation property implies that

$$d(dx_I) = d(dx_{i_1} \wedge \cdots \wedge dx_{i_k}) = \sum_{j=1}^k (-1)^{j-1} dx_{i_1} \wedge \cdots \wedge dx_{i_{j-1}} \wedge d(dx_{i_j}) \wedge dx_{i_{j+1}} \wedge \cdots \wedge dx_{i_k} = 0$$

since each $d(dx_{i_j}) = 0$. But then, again applying the anti-derivation condition together with linearity,

$$(d\omega)_p = d \left(\sum_I f_I dx_I \right) = \sum_I [(df_I) \wedge dx_I + f_I d(dx_I)] = \sum_I (df_I) \wedge dx_I.$$

Exercise 2.5.6. Go back to Section 2.2 and verify the various exterior derivative calculations we did there.

Proof of Proposition 2.5.4. Define d in local coordinates as in (2.3). Independence of the choice of coordinates will follow from uniqueness, so it remains to show:

- (a) This is an anti-derivation.
- (b) It satisfies the cocycle condition (ii).
- (c) Uniqueness.

(a) and (b) are both computations. Here's the computation for (b): If $\omega = \sum_I f_I dx_I$ in a neighborhood of $p \in M$, then

$$d\omega = \sum_I df_I \wedge dx_I = \sum_I \left(\sum_{i=1}^n \frac{\partial f}{\partial x_i} dx_i \right) \wedge dx_I$$

in that same neighborhood, so

$$d(d\omega) = \sum_I \left(\sum_{i=1}^n d \left(\frac{\partial f}{\partial x_i} \right) \wedge dx_I \right) \wedge dx_I = \sum_I \left(\sum_{i=1}^n \left(\sum_{j=1}^n \frac{\partial^2 f}{\partial x_j \partial x_i} dx_j \right) \wedge dx_I \right) \wedge dx_I = 0$$

since $\frac{\partial^2 f}{\partial x_j \partial x_i} = \frac{\partial^2 f}{\partial x_i \partial x_j}$ but $dx_j \wedge dx_i = -dx_i \wedge dx_j$.

Turning to (c), suppose there exists a degree +1 anti-derivation $D : \Omega^*(M) \rightarrow \Omega^*(M)$ satisfying (i) and (ii). In particular, $Df = df$ for all $f \in C^\infty(M) = \Omega^0(M)$. Then

$$D(dx_I) = D(dx_{i_1} \wedge \cdots \wedge dx_{i_k}) = \sum_{j=1}^k (-1)^{j-1} dx_{i_1} \wedge \cdots \wedge dx_{i_{j-1}} \wedge D(dx_{i_j}) \wedge dx_{i_{j+1}} \wedge \cdots \wedge dx_{i_k} = 0$$

since $D(dx_{i_j}) \stackrel{(i)}{=} D(Dx_{i_j}) \stackrel{(ii)}{=} 0$ by hypothesis.

Therefore, if $\omega = \sum_I f_I dx_I$ in a neighborhood of p , by linearity and the anti-derivation condition we have

$$D\omega = \sum_I (Df_I \wedge dx_I + f_I D(dx_I)) = \sum_I Df_I \wedge dx_I \stackrel{(i)}{=} \sum_I df_I \wedge dx_I = d\omega$$

in that neighborhood. Since this is true for all $p \in M$, it must be the case that $D\omega = d\omega$. \square

Notice that d being degree +1 and satisfying the cocycle condition (ii) says exactly that we have a diagram

$$\Omega^0(M) \xrightarrow{d_0} \Omega^1(M) \xrightarrow{d_1} \cdots \xrightarrow{d_{n-2}} \Omega^{n-1}(M) \xrightarrow{d_{n-1}} \Omega^n(M)$$

which is a cochain complex.

Definition 2.5.7. A differential form $\omega \in \Omega^k(M)$ is *closed* if $d\omega = 0$ and it is *exact* if $\omega = d\eta$ for some $\eta \in \Omega^{k-1}(M)$. Notice that, by the cocycle condition (ii), all exact forms are closed.

Exercise 2.5.8. Show that $\omega \in \Omega^1(\mathbb{R}^2 - \{0\})$ given by $\omega = \frac{x dy - y dx}{x^2 + y^2}$ is closed but not exact. (Using the correspondence between 1-forms and vector fields, this is equivalent to the fact that the vector field $\left(\frac{-y}{x^2 + y^2}, \frac{x}{x^2 + y^2} \right)$ is curl-free but not a gradient field.)

2.6 Pullbacks

Suppose we have manifolds M and N and a smooth map $f : M \rightarrow N$. Since we have a change of variables formula for integrals, and differential forms are supposed to be the objects that we can integrate on manifolds, we should expect that f gives a way of turning differential forms on one of M or N into forms on the other. If you think about how the change of variables formula works, you can figure out which direction this is supposed to go, but let's build up to this a bit more slowly.

First of all, recall that if $v \in T_p M$, then $df_p v \in T_{f(p)} N$; that is, $df_p : T_p M \rightarrow T_{f(p)} N$. More precisely, since $v = \alpha'(0)$ for some smooth curve α with $\alpha(0) = p$, we defined $df_p v = (f \circ \alpha)'(0)$. In words, f gives a way of pushing tangent vectors forward using df ,¹⁰ and we could have guessed this in advance because v

¹⁰In fact, many books and papers use the notation f_* rather than df since this is the standard notation for a pushforward operator.

is defined in terms of a map into M , which can then be composed with f to get a map into N . In categorical terms, tangent vectors and vector fields are transformed *covariantly*, meaning that tangent vectors or vector fields get sent in the same direction (from M to N) as the original map $f: M \rightarrow N$.

Now let's think about how we could move differential forms around using f . At a point, a k -form $\omega \in \Omega^k(M)$ corresponds to $\omega_p \in \wedge^k((T_p M)^*) \cong (\wedge^k(T_p M))^*$, or equivalently an alternating multilinear functional $(T_p M)^k \rightarrow \mathbb{R}$. In other words, unlike a tangent vector which is defined by a map *into* M , a differential form corresponds to a map *out of* $(T_p M)^k$. That means there's really no way to compose $\omega_p \in \wedge^k((T_p M)^*)$ with f to get something on N .

However, if we start instead with $\eta \in \Omega^k(N)$, then, as above, at a point $q \in N$, η_q corresponds to a map $(T_q N)^k \rightarrow \mathbb{R}$, so we *should* be able to precompose with a map $(T_p M)^k \rightarrow (T_q N)^k$ to get a k -form on M . More precisely:

Definition 2.6.1. If $f: M \rightarrow N$ is smooth and $\eta \in \Omega^k(N)$, then the *pullback* of η by f is the form $f^*\eta \in \Omega^k(M)$ defined by

$$(f^*\eta)_p(v_1, \dots, v_k) := \eta_{f(p)}(df_p v_1, \dots, df_p v_k)$$

for any $v_1, \dots, v_k \in T_p M$.

Notice, first of all, that this definition makes sense because $df_p: T_p M \rightarrow T_{f(p)} N$, so each $df_p v_i \in T_{f(p)} N$, and hence can be fed into $\eta_{f(p)}$.

In the paragraph before [Definition 2.6.1](#) I said we could precompose η with a map $(T_p M)^k \rightarrow (T_q N)^k$, but then didn't write down such a map. What's going on here? The differential $df_p: T_p M \rightarrow T_{f(p)} N$ induces a map $(df_p)^k: (T_p M)^k \rightarrow (T_{f(p)} N)^k$ by just applying df_p separately in each factor. So then another way of stating [Definition 2.6.1](#) is that

$$(f^*\eta)_p = \eta_{f(p)} \circ (df_p)^k.$$

Example 2.6.2. Suppose $g \in \Omega^0(N) = C^\infty(N)$. then

$$(f^*g)_p = g_{f(p)} = g(f(p)) = (g \circ f)(p),$$

recalling that, for a form ω , the notation ω_p really just means $\omega(p)$.

In other words, $f^*g = g \circ f$.

Example 2.6.3. Let $f: S^1 \rightarrow \mathbb{R}^2$ be the inclusion of the unit circle into the plane; i.e., $f(\theta) = (\cos \theta, \sin \theta)$. Let $\omega = dx \in \Omega^1(\mathbb{R}^2)$, which at each point $p \in \mathbb{R}^2$ is computed by $dx(a \frac{\partial}{\partial x} + b \frac{\partial}{\partial y}) = a$; or, said another way, by taking the dot product of the input tangent vector with the first standard basis vector.

What is $f^*\omega = f^*dx$? At any point $\theta_0 \in S^1$, we have the local coordinate basis $\{\frac{\partial}{\partial \theta}\}$ for $T_{\theta_0} S^1$, so we can completely specify f^*dx by determining the value of

$$(f^*dx)_{\theta_0} \left(a \frac{\partial}{\partial \theta} \right)$$

for any $a \in \mathbb{R}$. By [Definition 2.6.1](#),

$$(f^*dx)_{\theta_0} \left(a \frac{\partial}{\partial \theta} \right) = (dx)_{f(\theta_0)} \left(df_{\theta_0} \left(a \frac{\partial}{\partial \theta} \right) \right).$$

So first we'll need to compute $df_{\theta_0} \left(a \frac{\partial}{\partial \theta} \right)$. We can write $a \frac{\partial}{\partial \theta} = \alpha'(0)$, where $\alpha(t) = \theta_0 + at$, so [Definition 1.3.1](#) tells us that

$$\begin{aligned} df_{\theta_0} \left(a \frac{\partial}{\partial \theta} \right) &= (f \circ \alpha)'(0) = \frac{d}{dt} \Big|_{t=0} f(\alpha(t)) = \frac{d}{dt} \Big|_{t=0} f(\theta_0 + at) = \frac{d}{dt} \Big|_{t=0} (\cos(\theta_0 + at), \sin(\theta_0 + at)) \\ &= (-a \sin(\theta_0 + at), a \cos(\theta_0 + at))|_{t=0} = (-a \sin(\theta_0), a \cos(\theta_0)) = -a \sin(\theta_0) \frac{\partial}{\partial x} + a \cos(\theta_0) \frac{\partial}{\partial y} \end{aligned}$$

after switching between tuple notation and local coordinate basis notation.

Therefore,

$$(f^* dx)_{\theta_0} \left(a \frac{\partial}{\partial \theta} \right) = (dx)_{f(\theta_0)} \left(df_{\theta_0} \left(a \frac{\partial}{\partial \theta} \right) \right) = (dx)_{f(\theta_0)} \left(-a \sin(\theta_0) \frac{\partial}{\partial x} + a \cos(\theta_0) \frac{\partial}{\partial y} \right) = -a \sin(\theta_0).$$

In other words,

$$(f^* dx)_{\theta_0} = -\sin \theta_0 (d\theta)_{\theta_0}$$

or, by slight abuse of notation,

$$f^* dx = -\sin \theta d\theta.$$

Recalling that $\theta \in S^1$ maps to the point $(\cos \theta, \sin \theta)$, we can see that $\sin \theta$ is the y -coordinate, so this could also be written as

$$f^* dx = -y d\theta.$$

Geometrically, this makes sense: at the point $(x, y) = (\cos \theta, \sin \theta)$ this form is dual to the tangent vector $-y \frac{\partial}{\partial \theta}$, which at each point is just the orthogonal projection of the first standard basis vector $\frac{\partial}{\partial x}$ onto the tangent space to the circle, and of course $\frac{\partial}{\partial x}$ is the tangent vector to $(x, y) \in \mathbb{R}^2$ dual to dx . See [Figure 2.2](#).

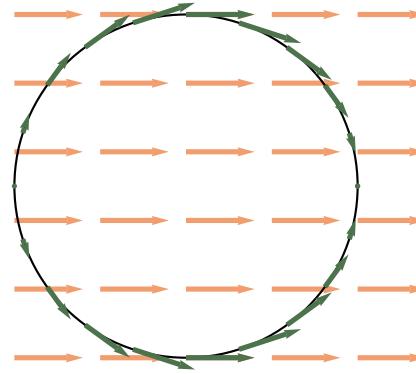


Figure 2.2: The vector field $\frac{\partial}{\partial x}$ (dual to dx) on \mathbb{R}^2 is shown in orange, and the vector field $-y \frac{\partial}{\partial \theta}$ (dual to $-y d\theta = f^* dx$) on the circle is shown in green.

One nice thing about pullbacks is that they commute with the exterior derivative:

Proposition 2.6.4. *For $f: M \rightarrow N$ smooth and $\omega \in \Omega^k(N)$,*

$$d(f^* \omega) = f^* d\omega.$$

Before proving this, let's see it in an example:

Example 2.6.5. Continuing Example 2.5.5, recall that $\omega = z dy + xy dz$ and we computed

$$d\omega = y dx \wedge dz + (x - 1)dy \wedge dz.$$

Now, consider the map $f: \mathbb{R}^2 \rightarrow \mathbb{R}^3$ given by $f(u, v) = (u^2, u, uv)$. For $p = (u_0, v_0) \in \mathbb{R}^2$ and $w = a \frac{\partial}{\partial u} + b \frac{\partial}{\partial v} \in T_p \mathbb{R}^2$ for $i = 1, 2$, we can write $w = \alpha'_i(0)$ where $\alpha_i(t) = p + tw = (u_0, v_0) + t(a, b)$, so

$$\begin{aligned} df_p(w) &= (f \circ \alpha_i)'(0) = \left. \frac{d}{dt} \right|_{t=0} ((u_0 + ta)^2, u_0 + ta, (u_0 + ta)(v_0 + tb)) \\ &= (2au_0, a, av_0 + bu_0) = 2au_0 \frac{\partial}{\partial x} + a \frac{\partial}{\partial y} + (av_0 + bu_0) \frac{\partial}{\partial z}, \end{aligned}$$

where I'm going back and forth between tuple notation (in \mathbb{R}^3 with the standard basis) and local coordinate basis notation (on $T_{f(p)} \mathbb{R}^3$ with the local coordinate basis [which of course is the same as the standard basis]).

Therefore,

$$\begin{aligned} (f^* \omega)_p(w) &= \omega_{f(p)}(df_p(w)) = \left(u_0 v_0 dy + u_0^3 dz \right) \left(2au_0 \frac{\partial}{\partial x} + a \frac{\partial}{\partial y} + (av_0 + bu_0) \frac{\partial}{\partial z} \right) \\ &= au_0 v_0 (1 + u_0^2) + bu_0^4. \end{aligned}$$

Since this is what the linear functional $(f^* \omega)_{(u_0, v_0)}$ does to the tangent vector $w = a \frac{\partial}{\partial u} + b \frac{\partial}{\partial v}$, we see that $(f^* \omega)_{(u_0, v_0)} = u_0 v_0 (1 + u_0^2) du + u_0^4 dv$ or, now thinking globally,

$$f^* \omega = uv(1 + u^2)du + u^4 dv.$$

Hence,

$$\begin{aligned} d(f^* \omega) &= \left(d(uv(1 + u^2)) \right) \wedge du + \left(d(u^4) \right) \wedge dv \\ &= \left(\frac{\partial}{\partial u} (uv(1 + u^2)) du + \frac{\partial}{\partial v} (uv(1 + u^2)) dv \right) \wedge du + \left(\frac{\partial}{\partial u} (u^4) du + \frac{\partial}{\partial v} (u^4) dv \right) \wedge dv \\ &= u(1 + u^2)dv \wedge du + 4u^3 du \wedge dv \\ &= (3u^3 - u)du \wedge dv. \end{aligned}$$

On the other hand,

$$\begin{aligned} (f^* d\omega)_p \left(\frac{\partial}{\partial u}, \frac{\partial}{\partial v} \right) &= (d\omega)_{f(p)} \left(df_p \left(\frac{\partial}{\partial u} \right), df_p \left(\frac{\partial}{\partial v} \right) \right) \\ &= \left(u_0 dx \wedge dz + (u_0^2 - 1)dy \wedge dz \right) \left(2u_0 \frac{\partial}{\partial x} + \frac{\partial}{\partial y} + v_0 \frac{\partial}{\partial z}, u_0 \frac{\partial}{\partial z} \right) \\ &= 3u_0^3 - u_0. \end{aligned}$$

Notice that $(3u_0^3 - u_0)du \wedge dv$ would produce the same output for this (or any other) input, so $(f^* d\omega)_{(u_0, v_0)} = (3u_0^3 - u_0)du \wedge dv$. In other words (now thinking globally)

$$f^* d\omega = (3u^3 - u)du \wedge dv = d(f^* \omega),$$

as predicted by Proposition 2.6.4.

Proof of Proposition 2.6.4. Let $p \in M$. If $\omega = g \in \Omega^0(N) = C^\infty(N)$, then $f^*g = g \circ f$, so $d(f^*g) = d(g \circ f)$. On the other hand, for any $v \in T_p M$,

$$(f^*dg)_p v = dg_{f(p)}(df_p v) = d(g \circ f)_p(v) = d(f^*g)_p(v),$$

where we used [Definition 2.6.1](#) for the first equality, the Chain Rule for the second, and [Example 2.6.2](#) for the third. This completes the proof in the $k = 0$ case.

Now, suppose $k > 0$ and that x_1, \dots, x_n are local coordinates in a neighborhood of $f(p) \in N$, meaning that $\omega \in \Omega^k(N)$ has the form $\omega_{f(p)} = \sum_I g_I dx_I$ at $f(p)$, so that

$$(d\omega)_{f(p)} = \sum_I dg_I \wedge dx_I.$$

But then

$$(f^*d\omega)_p = \sum_I (f^*dg_I) \wedge (f^*dx_I) = \sum_I d(g_I \circ f) \wedge f^*dx_{i_1} \wedge \cdots \wedge f^*dx_{i_k} = \sum_I d(g_I \circ f) \wedge d(x_{i_1} \circ f) \wedge \cdots \wedge d(x_{i_k} \circ f),$$

where we've repeatedly used both the $k = 0$ case and the fact that $f^*(\alpha \wedge \beta) = (f^*\alpha) \wedge (f^*\beta)$ (which follows from HW #2, Problem 1).

On the other hand,

$$(f^*\omega)_p = \sum_I (g_I \circ f) f^*dx_I = \sum_I (g_I \circ f) f^*dx_{i_1} \wedge \cdots \wedge f^*dx_{i_k} = \sum_I (g_I \circ f) d(x_{i_1} \circ f) \wedge \cdots \wedge d(x_{i_k} \circ f),$$

so

$$(df^*\omega)_p = \sum_I d(g_I \circ f) \wedge d(x_{i_1} \circ f) \wedge \cdots \wedge d(x_{i_k} \circ f) = (f^*d\omega)_p,$$

and we conclude that the proposition is also true for $k > 0$. \square

2.7 Lie Derivatives and Cartan's Magic Formula

Recall [Definition 1.7.4](#), the definition of the Lie derivative for vector fields:

$$\mathcal{L}_X Y := \lim_{t \rightarrow 0} \frac{(d\Phi_{-t})_{\Phi_t(p)} Y(\Phi_t(p)) - Y(p)}{t},$$

where Φ_t is the local flow for $X \in \mathfrak{X}(M)$. We saw in [Proposition 1.7.5](#) that the Lie derivative is equal to the Lie bracket $[X, Y]$.

The key idea in this derivative was to compare Y at $\Phi_t(p)$ to Y at p , and to do so we needed to move $Y(\Phi_t(p))$ from $T_{\Phi_t(p)} M$ to $T_p M$, which we did with the differential of the *negative* flow $d\Phi_{-t}$.

We can use the same idea to get a Lie derivative for differential forms, though we have to be careful about how we move a form $\omega \in \Omega^k(M)$ at a point $\Phi_t(p)$ to p : whereas vector fields push forward, differential forms pull back (as we've just seen in [Section 2.6](#)). So rather than pushing forward with $d\Phi_{-t}$, we will pull back by Φ_t^* :

Definition 2.7.1. Let $X \in \mathfrak{X}(M)$ and $\omega \in \Omega^k(M)$. Then the *Lie derivative* of ω with respect to X is defined (at a point $p \in M$) by

$$(\mathcal{L}_X \omega)_p := \lim_{t \rightarrow 0} \frac{\Phi_t^*(\omega_{\Phi_t(p)}) - \omega_p}{t} = \left. \frac{d}{dt} \right|_{t=0} (\Phi_t^*)_p (\omega_{\Phi_t(p)}).$$

Even more generally, we can combine pushforwards and pullbacks to get Lie derivatives of arbitrary tensor fields:

Definition 2.7.2. Let $X \in \mathfrak{X}(M)$, $a \in \mathcal{T}_r^0(M)$, and $\beta \in \mathcal{T}_0^s(M)$. Then the *Lie derivative* of the (r, s) -tensor field $a \otimes \beta$ is defined (at a point $p \in M$) by

$$(\mathcal{L}_X a \otimes \beta)_p := \frac{d}{dt} \Big|_{t=0} \left[(d\Phi_{-t})_{\Phi_t(p)} (a_{\Phi_t(p)}) \otimes (\Phi_t^*)_p (\beta_{\Phi_t(p)}) \right].$$

Extending linearly gives the Lie derivative with respect to X for arbitrary elements of $\mathcal{T}_r^s(M)$.

Lemma 2.7.3. For $f \in C^\infty(M) = \Omega^0(M)$ and $X \in \mathfrak{X}(M)$,

$$\mathcal{L}_X f = X(f).$$

Proof. By Definition 2.7.1 (first equality), Example 2.6.2 (second), and Definition 1.2.1 (third),

$$\mathcal{L}_X f = \frac{d}{dt} \Big|_{t=0} \Phi_t^*(f) = \frac{d}{dt} \Big|_{t=0} f \circ \Phi_t = X(f).$$

□

Example 2.7.4. If M is a manifold and g is a Riemannian metric on M , a vector field $X \in \mathfrak{X}(M)$ is called a *Killing field*¹¹ if $\mathcal{L}_X g = 0$. Intuitively, this means that the flow generated by X is an isometry: it preserves distances. If you read about general relativity, you will often see references to Killing fields, which encode the symmetries of spacetime.

Let's compute $\mathcal{L}_X g$ for some vector fields X on S^2 , where g is the standard Riemannian metric induced by the inner product on \mathbb{R}^3 . If we work in cylindrical coordinates, we have the local coordinate chart $\phi: (0, 2\pi) \rightarrow S^2$ given by

$$\phi(\theta, z) = \left(\sqrt{1 - z^2} \cos \theta, \sqrt{1 - z^2} \sin \theta, z \right).$$

So then our local coordinate basis $\{\frac{\partial}{\partial \theta}, \frac{\partial}{\partial z}\}$ at a point $p = \phi(\theta_0, z_0)$ looks like

$$\frac{\partial}{\partial \theta} \Big|_p = \frac{d}{dt} \Big|_{t=0} \phi(\theta_0 + t, z_0) = \left(-\sqrt{1 - z_0^2} \sin \theta_0, \sqrt{1 - z_0^2} \cos \theta_0, z_0 \right)$$

and

$$\frac{\partial}{\partial z} \Big|_p = \frac{d}{dt} \Big|_{t=0} \phi(\theta_0, z_0 + t) = \left(-\frac{z_0 \cos \theta_0}{\sqrt{1 - z_0^2}}, -\frac{z_0 \sin \theta_0}{\sqrt{1 - z_0^2}}, 1 \right).$$

See Figure 2.3 (which is the same as Figure 1.11).

¹¹Named after Wilhelm Killing.

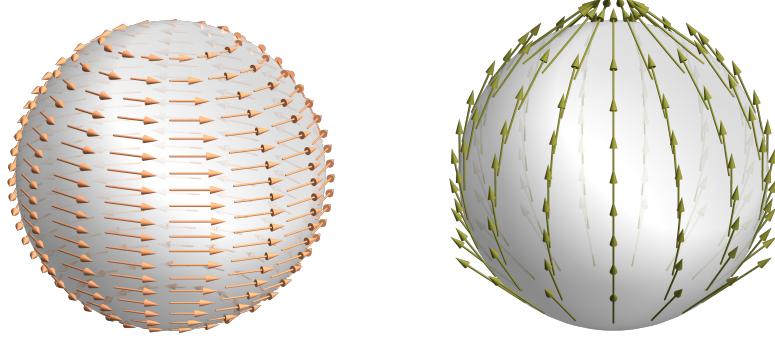


Figure 2.3: The vector fields $X = \frac{\partial}{\partial\theta}$ and $Y = \frac{\partial}{\partial z}$ on S^2 .

In turn, this implies that the Riemannian metric g is given by

$$\begin{aligned} g_p \left(a \frac{\partial}{\partial\theta} + b \frac{\partial}{\partial z}, c \frac{\partial}{\partial\theta} + d \frac{\partial}{\partial z} \right) &= \left[a \left(-\sqrt{1-z_0^2} \sin\theta_0, \sqrt{1-z_0^2} \cos\theta_0, z_0 \right) + b \left(-\frac{z_0 \cos\theta_0}{\sqrt{1-z_0^2}}, -\frac{z_0 \sin\theta_0}{\sqrt{1-z_0^2}}, 1 \right) \right] \\ &\quad \cdot \left[c \left(-\sqrt{1-z_0^2} \sin\theta_0, \sqrt{1-z_0^2} \cos\theta_0, z_0 \right) + d \left(-\frac{z_0 \cos\theta_0}{\sqrt{1-z_0^2}}, -\frac{z_0 \sin\theta_0}{\sqrt{1-z_0^2}}, 1 \right) \right] \\ &= (1-z_0^2) ac + \frac{1}{1-z_0^2} bd \end{aligned}$$

or, more symbolically,

$$g_{(\theta,z)} = (1-z^2) d\theta \otimes d\theta + \frac{1}{1-z^2} dz \otimes dz.$$

Notice that this only depends on z and not on θ , so we would expect that flowing by $\frac{\partial}{\partial\theta}$ should preserve this inner product; that is, $\frac{\partial}{\partial\theta}$ should be a Killing field.

Indeed, the local flow of $\frac{\partial}{\partial\theta}$ is given at a point $p = \phi(\theta_0, z_0)$ by

$$\Phi_t(p) = \phi(\theta_0 + t, z) = \left(\sqrt{1-z_0^2} \cos(\theta_0 + t), \sqrt{1-z_0^2} \sin(\theta_0 + t), z_0 \right),$$

Now, writing $\frac{\partial}{\partial \theta} = \alpha'(0)$ where $\alpha(s) = \phi(\theta_0 + s, z_0)$, we see that

$$\begin{aligned}
(d\Phi_t)_P \left(\frac{\partial}{\partial \theta} \Big|_P \right) &= (d\Phi_t)_P \alpha'(0) = \frac{d}{ds} \Big|_{s=0} (\Phi_t \circ \alpha)(s) \\
&= \frac{d}{ds} \Big|_{s=0} (\Phi_t(\phi(\theta_0 + s, z_0))) \\
&= \frac{d}{ds} \Big|_{s=0} \left(\sqrt{1 - z_0^2} \cos(\theta_0 + s + t), \sqrt{1 - z_0^2} \sin(\theta_0 + s + t), z_0 \right) \\
&= \left(-\sqrt{1 - z_0^2} \sin(\theta_0 + t), \sqrt{1 - z_0^2} \cos(\theta_0 + t), 0 \right) \\
&= \frac{\partial}{\partial \theta} \Big|_{\Phi_t(P)} ;
\end{aligned}$$

i.e., the local flow of $\frac{\partial}{\partial \theta}$ preserves $\frac{\partial}{\partial \theta}$, as you might expect.

On the other hand, $\frac{\partial}{\partial z} = \beta'(0)$ where $\beta(s) = \phi(\theta_0, z_0 + s)$, so

$$\begin{aligned}
(d\Phi_t)_P \left(\frac{\partial}{\partial z} \Big|_P \right) &= (d\Phi_t)_P \beta'(0) = \frac{d}{ds} \Big|_{s=0} (\Phi_t \circ \beta)(s) \\
&= \frac{d}{ds} \Big|_{s=0} (\Phi_t(\phi(\theta_0, z_0 + s))) \\
&= \frac{d}{ds} \Big|_{s=0} \left(\sqrt{1 - (z_0 + s)^2} \cos(\theta_0 + t), \sqrt{1 - (z_0 + s)^2} \sin(\theta_0 + t), z_0 + s \right) \\
&= \left(-\frac{z_0 \cos(\theta_0 + t)}{\sqrt{1 - z_0^2}}, -\frac{z_0 \sin(\theta_0 + t)}{\sqrt{1 - z_0^2}}, 1 \right) \\
&= \frac{\partial}{\partial z} \Big|_{\Phi_t(P)} ,
\end{aligned}$$

so flowing by $\frac{\partial}{\partial \theta}$ also preserves $\frac{\partial}{\partial z}$.

Therefore, for any $a \frac{\partial}{\partial \theta} \Big|_P + b \frac{\partial}{\partial z} \Big|_P \in T_P S^2$, we have

$$(d\Phi_t)_P \left(a \frac{\partial}{\partial \theta} \Big|_P + b \frac{\partial}{\partial z} \Big|_P \right) = a \frac{\partial}{\partial \theta} \Big|_{\Phi_t(P)} + b \frac{\partial}{\partial z} \Big|_{\Phi_t(P)} .$$

Since p and $\Phi_t(p)$ have the same z -coordinate,

$$\begin{aligned}
g_{\Phi_t(P)} \left(a \frac{\partial}{\partial \theta} \Big|_{\Phi_t(P)} + b \frac{\partial}{\partial z} \Big|_{\Phi_t(P)}, c \frac{\partial}{\partial \theta} \Big|_{\Phi_t(P)} + d \frac{\partial}{\partial z} \Big|_{\Phi_t(P)} \right) &= (1 - z_0^2)ac + \frac{1}{1 - z_0^2} bd \\
&= g_P \left(a \frac{\partial}{\partial \theta} \Big|_P + b \frac{\partial}{\partial z} \Big|_P, c \frac{\partial}{\partial \theta} \Big|_P + d \frac{\partial}{\partial z} \Big|_P \right) .
\end{aligned}$$

Hence, for any $u, v \in T_p S^2$

$$(\Phi_t^* g_{\Phi_t(p)})(u, v) = g_{\Phi_t(p)}((d\Phi_t)_p u, (d\Phi_t)_p v) = g_p(u, v)$$

is independent of t , and we see that

$$(\mathcal{L}_{\frac{\partial}{\partial \theta}} g)_p(u, v) = \left. \frac{d}{dt} \right|_{t=0} (\Phi_t^* g_{\Phi_t(p)})(u, v) = 0.$$

Therefore, $\frac{\partial}{\partial \theta}$ is a Killing field on the sphere, which makes sense: the local flow is just rotation around the z -axis, which certainly preserves distances.

On the other hand,

$$\Psi_t(p) = \phi(\theta_0, z_0 + t) = \left(\sqrt{1 - (z_0 + t)^2} \cos \theta_0, \sqrt{1 - (z_0 + t)^2} \sin \theta_0, z_0 + t \right)$$

is the local flow of $\frac{\partial}{\partial z}$. So then

$$\begin{aligned} (d\Psi_t)_p \left(\left. \frac{\partial}{\partial \theta} \right|_p \right) &= (d\Psi_t)_p \alpha'(0) = \left. \frac{d}{ds} \right|_{s=0} (\Psi_t \circ \alpha)(s) \\ &= \left. \frac{d}{ds} \right|_{s=0} (\Psi_t(\phi(\theta_0 + s, z_0))) \\ &= \left. \frac{d}{ds} \right|_{s=0} \left(\sqrt{1 - (z_0 + t)^2} \cos(\theta_0 + s), \sqrt{1 - (z_0 + t)^2} \sin(\theta_0 + s), z_0 + t \right) \\ &= \left(-\sqrt{1 - (z_0 + t)^2} \sin \theta_0, \sqrt{1 - (z_0 + t)^2} \cos \theta_0, 0 \right) \\ &= \left. \frac{\partial}{\partial \theta} \right|_{\Psi_t(p)} \end{aligned}$$

and

$$\begin{aligned} (d\Psi_t)_p \left(\left. \frac{\partial}{\partial z} \right|_p \right) &= (d\Psi_t)_p \beta'(0) = \left. \frac{d}{ds} \right|_{s=0} (\Psi_t \circ \beta)(s) \\ &= \left. \frac{d}{ds} \right|_{s=0} (\Psi_t(\phi(\theta_0, z_0 + s))) \\ &= \left. \frac{d}{ds} \right|_{s=0} \left(\sqrt{1 - (z_0 + s + t)^2} \cos \theta_0, \sqrt{1 - (z_0 + s + t)^2} \sin \theta_0, z_0 + s + t \right) \\ &= \left(-\frac{(z_0 + t) \cos \theta_0}{\sqrt{1 - (z_0 + t)^2}}, -\frac{(z_0 + t) \sin \theta_0}{\sqrt{1 - (z_0 + t)^2}}, 1 \right) \\ &= \left. \frac{\partial}{\partial z} \right|_{\Psi_t(p)}, \end{aligned}$$

so flowing by $\frac{\partial}{\partial z}$ also preserves the coordinate vector fields, but the coordinate vector fields do not have constant length on S^2 .

We see that

$$\begin{aligned} (\Psi_t^* g_{\Psi_t(p)}) \left(a \frac{\partial}{\partial \theta} \Big|_p + b \frac{\partial}{\partial z} \Big|_p, c \frac{\partial}{\partial \theta} \Big|_p + d \frac{\partial}{\partial z} \Big|_p \right) &= g_{\Psi_t(p)} \left(a \frac{\partial}{\partial \theta} \Big|_{\Psi_t(p)}, b \frac{\partial}{\partial z} \Big|_{\Psi_t(p)}, c \frac{\partial}{\partial \theta} \Big|_{\Psi_t(p)} + d \frac{\partial}{\partial z} \Big|_{\Psi_t(p)} \right) \\ &= (1 - (z_0 + t)^2)ac + \frac{1}{1 - (z_0 + t)^2}bd, \end{aligned}$$

and hence

$$\begin{aligned} (\mathcal{L}_{\frac{\partial}{\partial z}} g)_p \left(\frac{\partial}{\partial \theta} \Big|_p, \frac{\partial}{\partial \theta} \Big|_p \right) &= \frac{d}{dt} \Big|_{t=0} (\Psi_t^* g_{\Psi_t(p)}) \left(\frac{\partial}{\partial \theta} \Big|_p, \frac{\partial}{\partial \theta} \Big|_p \right) = \frac{d}{dt} \Big|_{t=0} \left[(1 - (z_0 + t)^2)ac + \frac{1}{1 - (z_0 + t)^2}bd \right] \\ &= -2z_0 ac + \frac{2z_0}{(1 - z_0^2)^2} bd; \end{aligned}$$

that is,

$$\mathcal{L}_{\frac{\partial}{\partial z}} g = -2z d\theta \otimes d\theta + \frac{2z}{(1 - z^2)^2} dz \otimes dz,$$

which is no longer positive-definite (and hence not a Riemannian metric), but is still a symmetric $(0, 2)$ -tensor field on S^2 which captures how g changes as we flow in the $\frac{\partial}{\partial z}$ direction.

One of the most important formulas in differential geometry is:

Theorem 2.7.5 (Cartan's Magic Formula). *If $\omega \in \Omega^k(M)$ and $X \in \mathfrak{X}(M)$, then*

$$\mathcal{L}_X \omega = \iota_X d\omega + d\iota_X \omega.$$

Here ι_X is the *contraction operator* (or *interior product*) defined as follows: for $\omega \in \Omega^k(M)$ and $X \in \mathfrak{X}(M)$, $\iota_X \omega$ is the $(k-1)$ -form defined by

$$(\iota_X \omega)_p(v_1, \dots, v_{k-1}) = \omega_p(X(p), v_1, \dots, v_{k-1})$$

for any $v_1, \dots, v_{k-1} \in T_p M$. This is sometimes also denoted by $X \lrcorner \omega$. This is an antiderivation of degree -1 .

The reason Cartan's magic formula is so important is that it is one of the most useful tools out there for computing exterior derivatives.

Example 2.7.6. Remember [Section 1.8](#), where we computed Lie brackets on S^3 . As a reminder, we had mutually perpendicular vector fields $X, Y, Z \in \mathfrak{X}(S^3)$ defined by

$$X(p) = pi, \quad Y(p) = pj, \quad Z(p) = pk$$

for all $p \in S^3$ thought of as unit quaternions, and showed that

$$[X, Y] = 2Z, \quad [Y, Z] = 2X, \quad [Z, X] = 2Y.$$

Let α, β, γ be the dual 1-forms; that is, $\alpha(X) = 1$, $\alpha(Y) = 0 = \alpha(Z)$ and similarly for β and γ . Then, using Cartan's magic formula,

$$d\alpha(Y, Z) = (\iota_Y d\alpha)(Z) = (\mathcal{L}_Y \alpha)(Z) - d(\iota_Y \alpha)(Z) = (\mathcal{L}_Y \alpha)(Z) - d(\alpha(Y))(Z) = (\mathcal{L}_Y \alpha)(Z)$$

since $\alpha(Y) = 0$.

The Lie derivative turns out to be a derivation, so it obeys a Leibniz rule:

$$\mathcal{L}_U(\omega(Y_1, \dots, Y_k)) = (\mathcal{L}_U\omega)(Y_1, \dots, Y_k) + \sum_{j=1}^k \omega(Y_1, \dots, \mathcal{L}_U Y_j, \dots, Y_k).$$

In particular,

$$0 = \mathcal{L}_Y(0) = \mathcal{L}_Y(\alpha(Z)) = (\mathcal{L}_Y\alpha)(Z) + \alpha(\mathcal{L}_Y Z) = (\mathcal{L}_Y\alpha)(Z) + \alpha([Y, Z]) = (\mathcal{L}_Y\alpha)(Z) + \alpha(2X) = (\mathcal{L}_Y\alpha)(Z) + 2,$$

so we conclude that

$$d\alpha(Y, Z) = (\mathcal{L}_Y\alpha)(Z) = -2.$$

By analogous calculations, we can show that $d\alpha(X, Y) = 0 = d\alpha(Z, X)$, so we conclude that

$$d\alpha = -2\beta \wedge \gamma,$$

and similarly one can show that $d\beta = 2\gamma \wedge \alpha$ and $d\gamma = -2\alpha \wedge \beta$.

This implies, in particular, that $\alpha \wedge d\alpha = -2\alpha \wedge \beta \wedge \gamma$ is never zero, so α is an example of what is called a *contact form* on S^3 .

Notice, in particular, that we never had to work in local coordinates in [Example 2.7.6](#), which is one of the great virtues of Cartan's magic formula: it is one of the few tools that allows one to compute without using coordinates.

Hopefully this gives you some reason to believe that Cartan's magic formula is useful, so let's try to prove it:

Proof of Theorem 2.7.5. If $u \in \Omega^0(M) = C^\infty(M)$, then $\mathcal{L}_X u = X(u)$ by [Lemma 2.7.3](#) and

$$\iota_X du + d\iota_X u = du(X) + 0 = du(X) = X(u) = \mathcal{L}_X u,$$

so the formula holds for 0-forms.

Now consider a 1-form ω which is the differential of a 0-form: $\omega = du$. If $Y \in \mathfrak{X}(M)$, then, by [Definition 2.7.1](#),

$$(\mathcal{L}_X \omega)_p(Y) = (\mathcal{L}_x du)_p(Y) = \frac{d}{dt} \Big|_{t=0} \Phi_t^*(du_{\Phi_t(p)})(Y) = (\mathcal{L}_x du)_p(Y) = \frac{d}{dt} \Big|_{t=0} du_{\Phi_t(p)}((d\Phi_t)_p(Y))$$

using the definition of pullback ([Definition 2.6.1](#)). In turn, the chain rule tells us that the right hand side is equal to

$$\frac{d}{dt} \Big|_{t=0} d(u \circ \Phi_t)_p(Y) = \frac{d}{dt} \Big|_{t=0} Y(u \circ \Phi_t)(p)$$

by [Lemma 1.3.4](#). Since Y is independent of t , the (1-dimensional) chain rule tell us that this is equal to

$$Y \left(\frac{d(u \circ \Phi_t)}{dt} \Big|_{t=0} \right) (p) = Y(X(u))(p).$$

So we have shown that $(\mathcal{L}_X \omega)_p(Y) = Y(X(u))(p)$.

On the other hand, $\iota_X d\omega = \iota_X d(du) = 0$ and

$$d(\iota_X \omega)(Y) = d(\iota_X du)(Y) = d(du(X))(Y) = d(X(U))(Y) = Y(X(U)) = (\mathcal{L}_X \omega)_p(Y)$$

by repeatedly using Lemma 1.3.4. Hence, the formula holds for differentials.

Finally, then, if $\omega \in \Omega^k(M)$ and we write $\omega = \sum_I a_I dx_I$ in local coordinates, then

$$\mathcal{L}_X \omega = \sum_I [(\mathcal{L}_X a_I) dx_I + a_I \mathcal{L}_X(dx_I)]. \quad (2.4)$$

The $k = 0$ argument above implies that

$$\mathcal{L}_X a_I = (\iota_X d + d\iota_X) a_I. \quad (2.5)$$

On the other hand, again using the fact that the Lie derivative is a derivation, we have

$$\begin{aligned} \mathcal{L}_X(dx_{i_1} \wedge \cdots \wedge dx_{i_k}) &= \sum_{j=1}^k dx_{i_1} \wedge \cdots \wedge \mathcal{L}_X dx_{i_j} \wedge \cdots \wedge dx_{i_k} = \sum_{j=1}^k dx_{i_1} \wedge \cdots \wedge (\iota_X d + d\iota_X) dx_{i_j} \wedge \cdots \wedge dx_{i_k} \\ &= (\iota_X d + d\iota_X)(dx_{i_1} \wedge \cdots \wedge dx_{i_k}) \end{aligned}$$

since each dx_{i_j} is a differential.

Combining this with (2.4) and (2.5) implies that

$$\mathcal{L}_X \omega = \sum_I [((\iota_X d + d\iota_X) a_I) dx_I + a_I (\iota_X d + d\iota_X)(dx_I)] = (\iota_X d + d\iota_X) \sum_I a_I dx_I = (\iota_X d + d\iota_X) \omega,$$

as desired. \square

2.8 De Rham Cohomology

Recall that we have the cochain complex

$$\Omega^0(M) \xrightarrow{d_0} \Omega^1(M) \xrightarrow{d_1} \cdots \xrightarrow{d_{n-2}} \Omega^{n-1}(M) \xrightarrow{d_{n-1}} \Omega^n(M)$$

where the subscript k indicates that d_k is the restriction of d to $\Omega^k(M)$.

Definition 2.8.1. The k th de Rham cohomology group of M is

$$H_{\text{dR}}^k(M) := \frac{\ker d_k}{\text{im } d_{k-1}}.$$

Example 2.8.2. We saw in Exercise 2.5.8 that the 1-form $\omega = \frac{x dy - y dx}{x^2 + y^2} \in \Omega^1(\mathbb{R}^2 - \{0\})$ is closed (in the kernel of d), but not exact (in the image of d), so it represents a non-trivial element of $H_{\text{dR}}^1(\mathbb{R}^2 - \{0\})$.

De Rham cohomology is a homotopy invariant of manifolds, meaning that homotopy equivalent manifolds have identical de Rham cohomology groups.

If $\omega \in \Omega^k(M)$ and $\eta \in \Omega^\ell(M)$ are closed (i.e., $d\omega = 0$ and $d\eta = 0$), meaning that they represent de Rham cohomology classes in $H_{\text{dR}}^k(M)$ and $H_{\text{dR}}^\ell(M)$, respectively, then

$$d(\omega \wedge \eta) = d\omega \wedge \eta + (-1)^k \omega \wedge d\eta = 0 \wedge \eta + (-1)^k \omega \wedge 0 = 0,$$

so $d(\omega \wedge \eta)$ is also closed, and hence represents a class in $H_{\text{dR}}^{k+\ell}(M)$. This means that

$$H_{\text{dR}}^*(M) := \bigoplus_k H_{\text{dR}}^k(M)$$

forms a (graded) ring, the *de Rham cohomology ring*, with the operation induced by \wedge .

Theorem 2.8.3 (de Rham theorem). *There exists a vector space isomorphism*

$$H_{\text{dR}}^k(M) \rightarrow H_k(M; \mathbb{R})^* \cong H^k(M; \mathbb{R}),$$

where $H_k(M; \mathbb{R})$ and $H^k(M; \mathbb{R})$ are the singular (or simplicial) homology and cohomology groups.

In fact, even more is true: the de Rham cohomology ring is isomorphic to the singular cohomology ring with real coefficients.

In Example 2.8.2, $H^1(\mathbb{R}^2 - \{0\}; \mathbb{R}) \cong H^1(S^1; \mathbb{R}) \cong \mathbb{R}$, so the 1-form $\omega = \frac{x dy - y dx}{x^2 + y^2} \in \Omega^1(\mathbb{R}^2 - \{0\})$ is a generator of $H_{\text{dR}}^1(\mathbb{R}^2 - \{0\})$.

We won't give all the details of the proof of de Rham's theorem, but the basic idea is that a closed form defines a linear functional on homology classes by integration. More precisely, the mapping $I: H_{\text{dR}}^k(M) \rightarrow H_k(M; \mathbb{R})^*$ is defined as follows: if $[\omega] \in H_{\text{dR}}^k(M)$ is represented by the closed form $\omega \in \Omega^k(M)$, then $I([\omega]) \in H_k(M; \mathbb{R})^*$ is given by

$$I([\omega])([C]) := \int_C \omega$$

for any $[C] \in H_k(M; \mathbb{R})$ which is represented by the cycle $C \subset M$. It's (hopefully!) clear that $I([\omega])$ is a linear functional (provided the integral here is anything sensible), but probably less obvious that I is well-defined and an isomorphism. Of course, to even make sense of the definition of I , we need to talk about integration.

2.9 Integration on Manifolds

As suggested in Section 2.1, the point of defining differential forms as alternating, multilinear gadgets is because they transform in the right way.

2.9.1 Digression on Differentials, Pullbacks, and Dual Maps

First, a quick refresher on *dual maps* or *algebraic adjoints*:

Given vector spaces U and V and a linear map $h: U \rightarrow V$, there is an associated dual map $h^*: V^* \rightarrow U^*$ defined as follows: if $\alpha \in V^*$, then $h^*(\alpha)$ is supposed to be an element of U^* ; that is, a linear functional on U , meaning that we can specify it by specifying what it does to an arbitrary element $u \in U$:

$$h^*(\alpha)(u) := \alpha(h(u)).$$

In other words, $h^*\alpha = \alpha \circ h$. It's easy to check that h^* is linear since h is.

Moreover, when U and V are finite-dimensional with chosen bases u_1, \dots, u_m for U and v_1, \dots, v_n for V , then the matrix for h^* with respect to the dual bases u_1^*, \dots, u_m^* for U^* and v_1^*, \dots, v_n^* for V^* is equal to the transpose (or conjugate transpose, for complex vector spaces) of the matrix for h with respect to the given bases for U and V . This explains the “algebraic adjoint” terminology, since the transpose (or conjugate transpose) is the usual adjoint, at least when the chosen bases are orthonormal. (And I claim it is occasionally useful to think about transpose in these terms, even though in applications we often think about the transpose as just a map $V \rightarrow U$.)

Now, consider a smooth map $g: M \rightarrow N$ and let $p \in M$. Then the differential $dg_p: T_p M \rightarrow T_{g(p)} N$ must have an associated dual map $(dg_p)^*: (T_{g(p)} N)^* \rightarrow (T_p M)^*$.

First of all, notice that $(T_{g(p)} N)^* = \wedge^1((T_{g(p)} N)^*)$ and $(T_p M)^* = \wedge^1((T_p M)^*)$, so the pullback (at p)

$$g_p^*: \wedge^1((T_{g(p)} N)^*) \rightarrow \wedge^1((T_p M)^*)$$

has the same domain and range as $(dg_p)^*$.

Moreover, by definition of the dual map, for a linear functional $\alpha \in (T_{g(p)} N)^*$ and a tangent vector $v \in T_p M$,

$$(dg_p)^*(\alpha)(v) = \alpha(dg_p(v)).$$

But this is exactly how we defined the pullback, where we now think of $\alpha = \omega_{g(p)}$, the value of the 1-form ω at $g(p)$.

In other words, g^* as defined in [Section 2.6](#) is the same as the dual map $(dg)^*$ associated to the differential dg .

2.9.2 How Differential Forms Transform

The goal is to see that transformations of differential forms under changes of coordinates are just like transformations of integrands under changes of coordinates.

As in [Section 2.1](#), let $g: B \rightarrow A$ be a diffeomorphism of open subsets of \mathbb{R}^n and let $f: A \rightarrow \mathbb{R}$ be a smooth function. Define

$$\omega = f dx_1 \wedge \cdots \wedge dx_n \in \Omega^n(A),$$

which pulls back to

$$g^* \omega = (f \circ g) g^*(dx_1) \wedge \cdots \wedge g^*(dx_n) = (f \circ g)(dg)^*(dx_1) \wedge \cdots \wedge (dg)^*(dx_n),$$

where here $(dg)^*$ means the dual of the differential dg . By Homework 2, Problem 1(c), the above is equal to

$$(f \circ g)(\det(dg)^*) dy_1 \wedge \cdots \wedge dy_n,$$

where we recall that y_1, \dots, y_n were just the names of the coordinates on B . Since determinants are invariant under taking transposes, $\det(dg)^* = \det dg = \det Jg$, so this parallels the usual vector calculus change of variables formula, at least when f is smooth.

In other words, differential forms transform in the right way to serve as integrands.

Example 2.9.1. Let $M = (0, +\infty) \times (-\pi, \pi)$, $N = \mathbb{R}^2$, and the map $g: M \rightarrow N$ given by

$$g(r, \theta) := (r \cos \theta, r \sin \theta).$$

In other words, this is the change of variables map between polar coordinates (r, θ) and Cartesian coordinates (x, y) on the plane.

Now, consider the standard area form $\omega = dx \wedge dy$ on N and we want to work out $g^*\omega$ from the definition.

By definition of the differential ([Definition 1.3.1](#)), we have that, for $p = (\theta_0, r_0) \in M$,

$$\begin{aligned} dg_p\left(\frac{\partial}{\partial \theta}\right) &= \frac{d}{dt}\Big|_{t=0} g(\theta_0 + t, r_0) = \frac{d}{dt}\Big|_{t=0} r_0(\cos(\theta_0 + t), \sin(\theta_0 + t)) \\ &= r_0(-\sin \theta_0, \cos \theta_0) = r_0\left(-\sin \theta_0 \frac{\partial}{\partial x} + \cos \theta_0 \frac{\partial}{\partial y}\right) \end{aligned}$$

where I move back and forth between tuple representations and local coordinate basis representations. Similarly,

$$\begin{aligned} dg_p\left(\frac{\partial}{\partial r}\right) &= \frac{d}{dt}\Big|_{t=0} g(\theta_0, r_0 + t) = \frac{d}{dt}\Big|_{t=0} (r_0 + t)(\cos \theta_0, \sin \theta_0) \\ &= (\cos \theta_0, \sin \theta_0) = \cos \theta_0 \frac{\partial}{\partial x} + \sin \theta_0 \frac{\partial}{\partial y}. \end{aligned}$$

Therefore,

$$\begin{aligned} (g^*\omega)_p\left(a \frac{\partial}{\partial \theta} + b \frac{\partial}{\partial r}, c \frac{\partial}{\partial \theta} + d \frac{\partial}{\partial r}\right) &= \omega_{g(p)}\left(dg_p\left(a \frac{\partial}{\partial \theta} + b \frac{\partial}{\partial r}\right), dg_p\left(c \frac{\partial}{\partial \theta} + d \frac{\partial}{\partial r}\right)\right) \\ &= dx \wedge dy \left((b \cos \theta_0 - ar_0 \sin \theta_0) \frac{\partial}{\partial x} + (b \sin \theta_0 + ar_0 \cos \theta_0) \frac{\partial}{\partial y}, \right. \\ &\quad \left. (d \cos \theta_0 - cr_0 \sin \theta_0) \frac{\partial}{\partial x} + (d \sin \theta_0 + cr_0 \cos \theta_0) \frac{\partial}{\partial y} \right) \\ &= (b \cos \theta_0 - ar_0 \sin \theta_0)(d \sin \theta_0 + cr_0 \cos \theta_0) \\ &\quad - (b \sin \theta_0 + ar_0 \cos \theta_0)(d \cos \theta_0 - cr_0 \sin \theta_0) \\ &= r(bc - ad). \end{aligned}$$

On the other hand,

$$dr \wedge d\theta \left(a \frac{\partial}{\partial \theta} + b \frac{\partial}{\partial r}, c \frac{\partial}{\partial \theta} + d \frac{\partial}{\partial r} \right) = bc - ad,$$

so we can conclude that $g^*\omega = r dr \wedge d\theta$.

Since the composition of the constant function 1 (on N) with g is just the constant function 1 (on M), the above discussion implies that

$$g^*\omega = g^*(dx \wedge dy) = (\det Jg)dr \wedge d\theta,$$

where $\det Jg$ is the Jacobian determinant

$$\begin{vmatrix} \frac{\partial g_1}{\partial r} & \frac{\partial g_1}{\partial \theta} \\ \frac{\partial g_2}{\partial r} & \frac{\partial g_2}{\partial \theta} \end{vmatrix} = \begin{vmatrix} \cos \theta & -r \sin \theta \\ \sin \theta & r \cos \theta \end{vmatrix} = r \cos^2 \theta + r \sin^2 \theta = r.$$

So the above discussion of how differential forms transform matches up to the standard vector calculus fact that the Cartesian area element $dx dy$ transforms to the polar area element $r dr d\theta$.

2.9.3 Definition of the Integral

First, we define integration of forms on open sets in \mathbb{R}^n : if $A \subset \mathbb{R}^n$ is open, then any $\omega \in \Omega^n(A)$ can be written as

$$\omega = f dx_1 \wedge \cdots \wedge dx_n,$$

where f is a smooth function on A . Define

$$\int_A \omega = \int_A f dx_1 \cdots dx_n,$$

where the integral on the right is just the usual Riemann integral of the (smooth) function f over A .

By the previous discussion, if $g : B \rightarrow A$ is an orientation-preserving diffeomorphism, then $\det Jg > 0$, so

$$\begin{aligned} \int_A \omega &= \int_A f dx_1 \cdots dx_n = \int_B (f \circ g) |\det Jg| dy_1 \cdots dy_n = \int_B (f \circ g)(\det Jg) dy_1 \cdots dy_n \\ &= \int_B (f \circ g)(\det Jg) dy_1 \wedge \cdots \wedge dy_n = \int_B g^* \omega. \end{aligned}$$

In other words, our definition is actually coordinate-independent, as you would hope.

If g is orientation-reversing, then we get

$$\int_A \omega = - \int_B g^* \omega.$$

Since we have no standard coordinates on an arbitrary manifold, these transformations are essential to defining integrals on manifolds.

Suppose next that we have an n -dimensional manifold M and the support of $\omega \in \Omega^n(M)$ is contained in a single coordinate chart (U, ϕ) so that $\phi : U \rightarrow \phi(U)$ is orientation-preserving. Then $\phi^* \omega \in \Omega^n(U)$, which we know how to integrate, so define

$$\int_M \omega = \int_U \phi^* \omega.$$

Proposition 2.9.2. *The above is well-defined.*

Proof. We need to check that, if the support of ω is also contained in some other coordinate chart (V, ψ) so that $\psi : V \rightarrow \psi(V)$ is orientation-preserving, then we get the same answer whether we compute $\int_M \omega = \int_U \phi^* \omega$ or $\int_M \omega = \int_V \psi^* \omega$.

We can move between U and V by way of the map $\psi^{-1} \circ \phi : U \rightarrow V$, which is orientation-preserving, so

$$\int_V \psi^* \omega = \int_U (\psi^{-1} \circ \phi)^* \psi^* \omega = \int_U \phi^* (\psi^{-1})^* \psi^* \omega = \int_U \phi^* (\psi \circ \psi^{-1})^* \omega = \int_U \phi^* \omega,$$

where I'm using the previously unstated fact that, for maps $f : A \rightarrow B$ and $g : B \rightarrow C$, $(g \circ f)^* = f^* g^*$, which follows from the chain rule. \square

So we've now succeeded in defining the integral of a form contained entirely in a single local coordinate chart. But what about arbitrary forms?

First of all, to avoid worrying about convergence issues, let's restrict to *compactly supported* forms: that is, those forms whose support (the set of points on which they are non-vanishing) is contained in some compact subset of M . Of course, if M is itself compact, this is no restriction at all.

So now suppose we have a compactly supported form $\omega \in \Omega^n(M)$. To define $\int_M \omega$, we will use a partition of unity argument.

Definition 2.9.3. Let T be a topological space and let $\{W_\alpha\}_{\alpha \in \mathcal{I}}$ be an open cover of T , where \mathcal{I} is some indexing set. A *partition of unity* subordinate to $\{W_\alpha\}_{\alpha \in \mathcal{I}}$ is a collection $\{\rho_\alpha\}_{\alpha \in \mathcal{I}}$ of continuous functions $\rho_\alpha : T \rightarrow [0, 1]$ so that $\text{supp}(\rho_\alpha) \subset W_\alpha$ (i.e., $\rho_\alpha(x) = 0$ for all $x \notin W_\alpha$) and, for all $x \in T$,

- there exists a neighborhood of x on which all but finitely many of the ρ_α are identically zero and
- $\sum_{\alpha \in \mathcal{I}} \rho_\alpha(x) = 1$ (note that the previous condition ensures this sum is finite and hence converges).

These always exist in reasonable topological spaces:

Theorem 2.9.4. Every open cover on a paracompact Hausdorff space has a subordinate partition of unity.

Since manifolds are second countable and Hausdorff, they are paracompact, so this implies (after modifying the proof to make the ρ_α smooth):

Corollary 2.9.5. Every manifold has a partition of unity subordinate to its coordinate atlas so that the functions are smooth.

Now, finally, we can define integration of compactly-supported forms on manifolds:

Definition 2.9.6. Let M be an n -dimensional manifold and let $\omega \in \Omega^n(M)$ be compactly supported. Let $\{\rho_\alpha\}$ be a partition of unity subordinate to the (maximal) atlas $\{(U_\alpha, \phi_\alpha)\}$ on M and define

$$\int_M \omega = \sum_{\alpha} \int_M \rho_\alpha \omega.$$

Remark 2.9.7. (i) $\rho_\alpha \omega$ is only non-vanishing inside the support of ρ_α , which is contained in some coordinate neighborhood $\phi_\alpha(U_\alpha)$, so the integrals inside the sum are well-defined.

- (ii) The local finiteness of the partition of unity guarantees only finitely many terms are nonzero on the compact support of ω , so the sum converges.
- (iii) If ω is already contained in some coordinate chart (ϕ, U) , then this definition agrees with the previous one: since $\sum_{\alpha} \rho_\alpha(p) = 1$ for each $p \in M$, we have that $\sum_{\alpha} \rho_\alpha \omega = \omega$, so using the previous definition

$$\int_M \omega = \int_U \phi^* \omega = \int_U \phi^* \left(\sum_{\alpha} \rho_\alpha \omega \right) = \sum_{\alpha} \int_U \phi^* (\rho_\alpha \omega) = \sum_{\alpha} \int_M \rho_\alpha \omega$$

by linearity of pullback and of integration on \mathbb{R}^n .

- (iv) If $\{\sigma_\alpha\}$ were a different subordinate partition of unity, then the previous observation implies that, or each α ,

$$\int_M \rho_\alpha \omega = \sum_\beta \int_M \sigma_\beta \rho_\alpha \omega$$

and similarly for each β

$$\int_M \sigma_\beta \omega = \sum_\alpha \int_M \rho_\alpha \sigma_\beta \omega,$$

so

$$\sum_\alpha \int_M \rho_\alpha \omega = \sum_\alpha \sum_\beta \int_M \sigma_\beta \rho_\alpha \omega = \sum_\beta \sum_\alpha \int_M \rho_\alpha \sigma_\beta \omega = \sum_\beta \int_M \sigma_\beta \omega,$$

which means we get the same value for $\int_M \omega$ no matter which partition of unity we use.

- (v) Integration is a linear operator: for $a \in \mathbb{R}$ and $\omega, \eta \in \Omega^n(M)$, $\int_M (\omega + a\eta) = \int_M \omega + a \int_M \eta$.

- (vi) If $\omega \in \Omega^n(M)$ and $g : N \rightarrow M$ is an orientation-preserving diffeomorphism, then

$$\int_M \omega = \int_N g^* \omega.$$

Example 2.9.8. Let $\gamma : [a, b] \rightarrow M$ be a smooth curve with $\gamma(a) = p$ and $\gamma(b) = q$. Assume $f \in C^\infty(M)$ and $\omega = df \in \Omega^1(M)$. Then I claim that

$$\int_{[a,b]} \gamma^* \omega = f(q) - f(p).$$

This is just a computation:

$$\begin{aligned} \int_{[a,b]} \gamma^* \omega &= \int_{[a,b]} \gamma^*(df) = \int_{[a,b]} d(\gamma^* f) = \int_{[a,b]} d(f \circ \gamma) = \int_{[a,b]} (f \circ \gamma)'(t) dt \\ &= (f \circ \gamma)(b) - (f \circ \gamma)(a) = f(q) - f(p) \end{aligned}$$

where the second equality follows from [Proposition 2.6.4](#), the third from [Example 2.6.2](#), and the fifth from the fundamental theorem of calculus.

Example 2.9.9. Suppose $\gamma : S^1 \rightarrow M$ is smooth and $\omega \in \Omega^1(M)$. Then we define the *line integral* of ω around γ to be

$$\oint_\gamma \omega := \int_{S^1} \gamma^* \omega.$$

Now suppose $M = \mathbb{R}^n$, so that $\omega = a_1 dx_1 + \cdots + a_n dx_n$. Letting $\{\frac{\partial}{\partial \theta}\}$ be the local coordinate basis for the tangent space $T_t S^1$ at a point $t \in S^1$ and writing $\gamma(t) = (\gamma_1(t), \dots, \gamma_n(t))$, we can compute

$$(\gamma^* dx_i)_t \left(b \frac{\partial}{\partial \theta} \right) = (dx_i)_{\gamma(t)} \left(d\gamma_t \left(b \frac{\partial}{\partial \theta} \right) \right) = b(dx_i)_{\gamma(t)} \left(d\gamma_t \left(\frac{\partial}{\partial \theta} \right) \right) = b(dx_i)_{\gamma(t)} \left(\frac{\partial}{\partial \theta} (\gamma)(t) \right)$$

using [Lemma 1.3.4](#). But then

$$\frac{\partial}{\partial \theta} (\gamma)(t) = \gamma'_1(t) \frac{\partial}{\partial x_1} + \cdots + \gamma'_n(t) \frac{\partial}{\partial x_n},$$

so we see that

$$(\gamma^* dx_i)_t \left(b \frac{\partial}{\partial \theta} \right) = b \gamma'_i(t)$$

or, in other words, $(\gamma^* dx_i)_t = \gamma'_i(\theta) d\theta$. Therefore,

$$(\gamma^* \omega)_t = (\gamma^* a_1)_t (\gamma^* dx_1)_t + \cdots + (\gamma^* a_n)_t (\gamma^* dx_n)_t = [a_1(\gamma(t)) \gamma'_1(t) + \cdots + a_n(\gamma(t)) \gamma'_n(t)] d\theta.$$

Notice that the coefficient is just $A(\gamma(t)) \cdot \gamma'(t)$, where the vector field $A = (a_1, \dots, a_n)$ is dual to ω . Of course, this is the sort of expression you usually see in the Multivariable Calculus definition of a line integral.

Putting Examples 2.9.8 and 2.9.9 together, notice that if $\omega = df \in \Omega^1(M)$, then $\oint_{\gamma} \omega = 0$ for all closed curves γ in M .

Proposition 2.9.10. *The converse is true. That is, any $\omega \in \Omega^1(M)$ with the property that $\oint_{\gamma} \omega = 0$ for all closed curves γ in M is exact.*

Exercise 2.9.11. Prove this.

2.10 Integration on Submanifolds

So far, we've only defined integrating an n -form over an n -dimensional manifold. However, notice that any open subset of an n -dimensional manifold is also an n -manifold, so we can actually integrate over arbitrary open subsets.

Indeed, given an n -manifold M and any $\omega \in \Omega^n(M)$, we get a (signed) measure m on open subsets of M defined by

$$m(A) := \int_A \omega.$$

When ω is a volume form (meaning it is nowhere-vanishing), then $m(A)$ will have the same sign for all open A , and if the sign is positive then m will satisfy the axioms of a measure on the σ -algebra of Borel sets on M ; that is, it is a Borel measure. Moreover, if M is compact then $\int_M \omega$ is finite, so m can be normalized to give a probability measure. Indeed, starting with a volume form is by far the most common way of defining probability measures on (orientable) manifolds.

However, this is not so useful for the integrals mentioned in Section 2.8 that show up in de Rham cohomology. After all, if N is a k -dimensional submanifold of the n -dimensional manifold M with $k < n$ and $\omega \in \Omega^n(M)$, then $\int_N \omega = 0$.¹²

Rather, for de Rham cohomology purposes, we need a way of defining $\int_N \eta$ where $\eta \in \Omega^k(M)$. Intuitively, we want to “restrict” η to N here, but trying to write this intuition out directly leads to something quite complicated. But this is a place where all our effort in defining things in such a relatively abstract way pays off by giving us a simple and natural (though seemingly slightly indirect) way of defining our integral:

Definition 2.10.1. Suppose N is a k -dimensional submanifold of the manifold M and let $i: N \hookrightarrow M$ be the inclusion map. If $\eta \in \Omega^k(M)$, define

$$\int_N \eta := \int_N i^* \eta.$$

¹²As written, this doesn't quite make sense given our definitions, but one way to formalize it would be to realize N as the intersection of a decreasing nested sequence of open sets and then taking the limit of the integral of ω over those open sets. The measures of these sets will go to zero.

The point here is that $i^*\eta \in \Omega^k(N)$, and [Definition 2.9.6](#) applies.

Example 2.10.2. Let $M = \mathbb{R}^3$ and let S be a surface which is the graph of a function; i.e.,

$$S = \{(u, v, g(u, v)) : u, v \in \mathbb{R}\},$$

for some smooth function $g : \mathbb{R}^2 \rightarrow \mathbb{R}$ (this is not really a restrictive assumption: the implicit function theorem implies that locally every surface looks like this, possibly after permuting coordinates). We can parametrize S by the map $h : \mathbb{R}^2 \rightarrow \mathbb{R}^3$ given by $h(u, v) = (u, v, g(u, v))$.

Suppose $\omega \in \Omega^2(\mathbb{R}^3)$ is compactly supported. We can write

$$\omega = f_1 dy \wedge dz + f_2 dz \wedge dx + f_3 dx \wedge dy.$$

Then

$$\int_S \omega = \int_{\mathbb{R}^2} h^* \omega,$$

so we can compute

$$\begin{aligned} h^* dx &= d(x \circ h) = du \\ h^* dy &= d(y \circ h) = dv \\ h^* dz &= d(z \circ h) = d(g(u, v)) = \frac{\partial g}{\partial u} du + \frac{\partial g}{\partial v} dv, \end{aligned}$$

and hence

$$\begin{aligned} h^*(dx \wedge dy) &= h^* dx \wedge h^* dy = du \wedge dv \\ h^*(dy \wedge dz) &= h^* dy \wedge h^* dz = dv \wedge \left(\frac{\partial g}{\partial u} du + \frac{\partial g}{\partial v} dv \right) = -\frac{\partial g}{\partial u} du \wedge dv \\ h^*(dz \wedge dx) &= h^* dz \wedge h^* dx = \left(\frac{\partial g}{\partial u} du + \frac{\partial g}{\partial v} dv \right) \wedge du = -\frac{\partial g}{\partial v} du \wedge dv. \end{aligned}$$

Therefore,

$$\int_S \omega = \int_{\mathbb{R}^2} h^* \omega = \int_{\mathbb{R}^2} \left(-\frac{\partial g}{\partial u} f_1 - \frac{\partial g}{\partial v} f_2 + f_3 \right) du \wedge dv. \quad (2.6)$$

Now, I claim this is a standard formula from vector calculus. To see this, let $\vec{F} = (f_1, f_2, f_3)$ be the vector field corresponding to ω (in the notation from [Section 2.2](#), $\omega = \star F^\flat$) and let $\vec{n} = \left(-\frac{\partial g}{\partial u}, -\frac{\partial g}{\partial v}, 1 \right)$. Notice that, at $p = (u, v, g(u, v)) \in S$, the tangent space $T_p S$ is spanned by

$$\frac{d}{dt} \Big|_{t=0} h(t, 0) = \frac{d}{dt} \Big|_{t=0} (t, 0, g(t, 0)) = \left(1, 0, \frac{\partial g}{\partial u} \right)$$

and

$$\frac{d}{dt} \Big|_{t=0} h(0, t) = \frac{d}{dt} \Big|_{t=0} (0, t, g(0, t)) = \left(0, 1, \frac{\partial g}{\partial v} \right).$$

Since

$$\vec{n} \cdot \left(1, 0, \frac{\partial g}{\partial u} \right) = \left(-\frac{\partial g}{\partial u}, -\frac{\partial g}{\partial v}, 1 \right) \cdot \left(1, 0, \frac{\partial g}{\partial u} \right) = -\frac{\partial g}{\partial u} + \frac{\partial g}{\partial u} = 0$$

and

$$\vec{n} \cdot \left(0, 1, \frac{\partial g}{\partial v} \right) = \left(-\frac{\partial g}{\partial u}, -\frac{\partial g}{\partial v}, 1 \right) \cdot \left(0, 1, \frac{\partial g}{\partial v} \right) = -\frac{\partial g}{\partial v} + \frac{\partial g}{\partial v} = 0,$$

the vector \vec{n} is normal to S at all points of S .

Therefore, if $\vec{u} = \frac{\vec{n}}{\|\vec{n}\|}$ and $dA = \|\vec{n}\|du \wedge dv$ (usually called the area form or area element for S), then our formula (2.6) can be re-written as

$$\int_S \omega = \int_{\mathbb{R}^2} (\vec{F} \cdot \vec{u}) dA,$$

which is the standard formula from vector calculus for computing the surface integral of a vector field.

Example 2.10.3. Recall the form $\omega = \frac{x dy - y dx}{x^2 + y^2} \in \Omega^1(\mathbb{R}^2 - \{0\})$ defined in [Exercise 2.5.8](#) and also considered in [Example 2.8.2](#). Let $f : S^1 \rightarrow \mathbb{R}^2 - \{0\}$ defined by $f(\theta) = (\cos \theta, \sin \theta)$ be the standard parametrization of the unit circle. Then

$$\int_{S^1} \omega = \int_{S^1} f^* \omega.$$

We saw in [Example 2.6.3](#) that $f^* dx = -\sin \theta d\theta$ and a similar computation shows that $f^* dy = \cos \theta d\theta$. Since

$$\omega_{f(\theta_0)} = \frac{\cos(\theta_0) dy - \sin(\theta_0) dx}{\cos^2 \theta_0 + \sin^2 \theta_0} = \cos(\theta_0) dy - \sin(\theta_0) dx,$$

we see that

$$(f^* \omega)_{\theta_0} = (\cos^2 \theta_0 + \sin^2 \theta_0) d\theta = d\theta,$$

so

$$\int_{S^1} \omega = \int_{S^1} f^* \omega = \int_{S^1} d\theta = 2\pi.$$

2.11 Stokes' Theorem

At this point we have all the ingredients in place to prove (a version of) de Rham's theorem: for a k -dimensional homology class $[C] \in H_k(M; \mathbb{R})$ represented by a submanifold C , and a de Rham cohomology class $[\omega]$ represented by a closed form $\omega \in \Omega^k(M)$, we at least know what

$$I([\omega])([C]) := \int_C \omega$$

means.

However, to see that this is well-defined, we need to know that, if $[\omega_1] = [\omega_2]$, then $\int_C \omega_1 = \int_C \omega_2$, and similarly that if $[C_1] = [C_2]$, then $\int_{C_1} \omega = \int_{C_2} \omega$.

For the first, suppose $\omega_1, \omega_2 \in \Omega^k(M)$ represent the same de Rham cohomology class (i.e., the same element of $H_{dR}^k(M) = \frac{\ker d}{\text{im } d}$), then $\omega_1 - \omega_2 = d\eta$ for some $\eta \in \Omega^{k-1}(M)$. So we need to show that

$$0 = \int_C d\eta = \int_C (\omega_1 - \omega_2) = \int_C \omega_1 - \int_C \omega_2.$$

Since C is a cycle, $\partial C = \emptyset$, so this will follow from:

Theorem 2.11.1 (Stokes' Theorem for Differential Forms). *If M is a compact, oriented n -manifold with boundary and $\omega \in \Omega^{n-1}(M)$, then*

$$\int_M d\omega = \int_{\partial M} \omega.$$

As for the second equality we need to show that the pairing I is well-defined, suppose $[C_1] = [C_2]$. This means¹³ that $\partial E = C_2 - C_1$ for some $(k+1)$ -chain E . But then for any closed $\omega \in \Omega^k(M)$,

$$\int_{C_2} \omega - \int_{C_1} \omega = \int_{C_2 - C_1} \omega = \int_E d\omega = \int_E 0 = 0$$

by [Theorem 2.11.1](#).

In other words, Stokes' Theorem is the key to showing that the pairing I is well-defined. Of course, to really properly prove [Theorem 2.8.3](#) we would also need to argue that every homology class can be represented by a submanifold (or at least a finite union of submanifolds), which we're not going to bother with.

To give a complete proof of [Theorem 2.11.1](#), we would need to develop the theory of manifolds with boundary. However, since we won't really be using that theory for anything else, I don't think it's worth the effort. Instead, I'll just give the part of the proof that doesn't interact with the boundary. See Guillemin and Pollack [[15](#)] for details on manifolds with boundary and a complete proof of Stokes' Theorem.

First part of proof of Theorem 2.11.1. Notice, first of all, that by linearity of the integral we can assume ω has compact support contained in the image of some local coordinate chart (ϕ, U) , so that $\int_M d\omega = \int_U \phi^* d\omega = \int_U d\phi^* \omega$. If we assume $\phi(U) \cap \partial M = \emptyset$, then $\int_{\partial M} \omega = 0$, so the goal is to show that $\int_U d\phi^* \omega = 0$.

We know that $\phi^* \omega \in \Omega^{n-1}(U) \subset \Omega^{n-1}(\mathbb{R}^n)$, so we can write

$$\phi^* \omega = \sum_{i=1}^n (-1)^{i-1} f_i dx_1 \wedge \cdots \wedge d\check{x}_i \wedge \cdots \wedge dx_n$$

for some compactly supported functions f_i , where $d\check{x}_i$ means that dx_i is omitted and the signs are chosen so that

$$d\phi^* \omega = \left(\sum_{i=1}^n \frac{\partial f_i}{\partial x_i} \right) dx_1 \wedge \cdots \wedge dx_n.$$

Since the f_i are compactly supported, we can extend their domain of definition to all of \mathbb{R}^n by defining them to be zero outside U . Therefore,

$$\int_U d\phi^* \omega = \int_U \left(\sum_{i=1}^n \frac{\partial f_i}{\partial x_i} \right) dx_1 \wedge \cdots \wedge dx_n = \int_{\mathbb{R}^n} \left(\sum_{i=1}^n \frac{\partial f_i}{\partial x_i} \right) dx_1 \wedge \cdots \wedge dx_n = \sum_{i=1}^n \int_{\mathbb{R}^n} \frac{\partial f_i}{\partial x_i} dx_1 \cdots dx_n.$$

By Fubini's Theorem, we can interpret each term as an iterated integral and do the integration in any order we like:

$$\int_{\mathbb{R}^n} \frac{\partial f_i}{\partial x_i} dx_1 \cdots dx_n = \int_{\mathbb{R}^{n-1}} \left(\int_{-\infty}^{\infty} \frac{\partial f_i}{\partial x_i} dx_i \right) dx_1 \cdots d\check{x}_i \cdots dx_n.$$

¹³If you haven't seen homology before, you might just need to trust me on this.

Now we're thinking of the integrand $\int_{-\infty}^{\infty} \frac{\partial f_i}{\partial x_i} dx_i$ as a function of $(x_1, \dots, \check{x}_i, \dots, x_n)$. Which function? Well, to any specific point $(a_1, \dots, \check{a}_i, \dots, a_n)$ it assigns the number $\int_{-\infty}^{\infty} g'(t) dt$ where $g(t) = f_i(a_1, \dots, t, \dots, a_n)$. But since f_i is compactly supported, we know there exists T big enough that $g(t) = 0$ for all $t \geq T$.

Then the Fundamental Theorem of Calculus implies that

$$\int_{-\infty}^{\infty} g'(t) dt = \int_{-T}^T g'(t) dt = g(T) - g(-T) = 0 - 0 = 0,$$

so we have

$$\int_M d\omega = \int_U \phi^* d\omega = \int_U d\phi^* \omega = 0$$

as desired. \square

To give the rest of the proof of [Theorem 2.11.1](#), one needs to deal with the case $\phi(U) \cap \partial M \neq \emptyset$. Again, see Guillemin and Pollack [15] if you're interested.

Example 2.11.2. Suppose S is a smooth surface in \mathbb{R}^3 with connected boundary $\partial S = \Gamma$, which is parametrized by $\gamma: S^1 \rightarrow \mathbb{R}^3$. Let $\omega \in \Omega^1(\mathbb{R}^3)$. Then [Theorem 2.11.1](#) tells us that

$$\int_S d\omega = \int_{\Gamma} \omega. \quad (2.7)$$

Let's translate this into vector calculus language. Let $\vec{F} \in \mathfrak{X}(\mathbb{R}^3)$ be the vector field corresponding to ω . That is, if $\omega = f dx + g dy + h dz$, then $\vec{F} = f \frac{\partial}{\partial x} + g \frac{\partial}{\partial y} + h \frac{\partial}{\partial z}$.

Following [Example 2.9.9](#),

$$\int_{\Gamma} \omega = \int_{S^1} \gamma^* \omega = \int_{S^1} \vec{F} \cdot \gamma'(t) dt$$

which we often write as $\oint_{\Gamma} \vec{F} \cdot d\Gamma$.

To translate the left-hand side of (2.7), recall that we saw in [Section 2.2](#) that the vector field corresponding to $d\omega$ is $\nabla \times \vec{F}$. Then, following [Example 2.10.2](#) we have

$$\int_S d\omega = \int_{\mathbb{R}^2} ((\nabla \times \vec{F}) \cdot u) dA,$$

which is what we mean in vector calculus by the notation $\iint_S (\nabla \times \vec{F}) \cdot dS$, where dS is the area element on S .

In other words, translating the Stokes' theorem result (2.7) into vector calculus language recovers the vector calculus Stokes' Theorem (or Curl Theorem)

$$\iint_S (\nabla \times \vec{F}) \cdot dS = \oint_{\Gamma} \vec{F} \cdot d\Gamma.$$

Exercise 2.11.3. Do the same sort of translation for the Divergence Theorem.

Chapter 3

Lie Groups and Homogeneous Spaces

3.1 Some Motivation from de Rham Cohomology

Recall that we have the de Rham cohomology group $H_{\text{dR}}^k(M)$, which tells us a lot about the topology of the manifold, but can in general be quite challenging to compute. After all, the $\Omega^k(M)$ are $C^\infty(M)$ -modules, and hence infinite-dimensional as real vector spaces. If M is, e.g., compact then $H_{\text{dR}}^k(M)$ will be a finite-dimensional real vector space, but it appears as the quotient of two infinite-dimensional vector spaces, so it can be somewhat challenging to compute.

However, when the manifold has lots of symmetry, computing the de Rham cohomology groups can actually become a finite-dimensional problem. In this case, a reasonable interpretation of “lots of symmetry” is “admits a transitive action of a Lie group,” resulting in a class of manifolds called *homogeneous spaces*. We haven’t defined Lie groups yet (though we will shortly), but the idea is that it’s a group which is also a manifold in a sensible way. At a more conceptual level, it’s just a continuous family of symmetries of the manifold, where here “continuous” means they form a continuum.

The following statement says more precisely how de Rham cohomology becomes finite-dimensional in this setting, though don’t worry too much about terms in the statement that we haven’t defined yet.

Theorem 3.1.1. *Let M be a homogeneous space where the transitive action is by a compact, connected Lie group. Then:*

- *Each closed k -form on M is in the same de Rham cohomology class as a G -invariant closed k -form.*
- *If a G -invariant closed k -form is exact, then it is also the exterior derivative of some G -invariant $(k-1)$ -form.*

Corollary 3.1.2. *The cohomology of G -invariant forms on M is isomorphic to its de Rham cohomology.*

Since, as we will see, G -invariant forms are determined by their behavior at a single point, this means that the space of G -invariant forms will be finite-dimensional (it can’t be bigger than $\wedge^k(T_p M)^*$), and hence computing de Rham cohomology becomes a finite-dimensional linear algebra problem.

Of course, the simplest possible example of a homogeneous space is a manifold that is itself a group, which acts by itself by (e.g. left) multiplication.

Example 3.1.3. S^3 is a group: it's just the unit quaternions, and hence it has a group operation given by multiplication of quaternions. Moreover, the vector fields $X, Y, Z \in \mathfrak{X}(S^3)$ from [Section 1.8](#) given by

$$X(p) = pi, \quad Y(p) = pj, \quad Z(p) = pk$$

are invariant under left-multiplication, and hence the dual 1-forms $\alpha, \beta, \gamma \in \Omega^1(S^3)$ from [Example 2.7.6](#) are also invariant under left-multiplication. Moreover, it is easy to show that *all* left-invariant 1-forms must be of the form $a\alpha + b\beta + c\gamma$ for constants $a, b, c \in \mathbb{R}$, so the space of left-invariant 1-forms is 3-dimensional.

Likewise, $\alpha \wedge \beta, \beta \wedge \gamma, \gamma \wedge \alpha$ form a basis for the space of left-invariant 2-forms on S^3 , which is therefore also 3-dimensional.

We saw in [Example 2.7.6](#) that

$$d\alpha = -2\beta \wedge \gamma, \quad d\beta = -2\gamma \wedge \alpha, \quad d\gamma = -2\alpha \wedge \beta,$$

so there are *no* closed left-invariant 1-forms, and [Corollary 3.1.2](#) implies $H_{\text{dR}}^1(S^3) = \{0\}$.

On the other hand,

$$d(\beta \wedge \gamma) = d\beta \wedge \gamma - \beta \wedge d\gamma = (-2\gamma \wedge \alpha) \wedge \gamma - \beta \wedge (-2\alpha \wedge \beta) = 0 - 0 = 0$$

and likewise for the other left-invariant 2-forms, so the left-invariant 2-forms are all closed... but they're also all exact: for example, $\beta \wedge \gamma = d\left(-\frac{1}{2}\alpha\right)$. So this tells us that $H_{\text{dR}}^2(S^3) = \{0\}$.

Therefore, the cohomology of left-invariant forms on S^3 is simply

$$\begin{aligned} H_{\text{dR}}^0(S^3) &\cong \mathbb{R} \\ H_{\text{dR}}^1(S^3) &= \{0\} \\ H_{\text{dR}}^2(S^3) &= \{0\} \\ H_{\text{dR}}^3(S^3) &\cong \mathbb{R}, \end{aligned}$$

which agrees with the usual (simplicial/singular/de Rham/sheaf/whatever) cohomology of S^3 .

The point here is that de Rham cohomology simplifies in the presence of symmetry.

Of course, this is a general principle: typically we expect lots of computations to simplify in the presence of symmetry. Lie groups provide the proper notion of (continuous) symmetries of manifolds, so if we want to understand symmetries and use them to simplify problems, we should probably care at least a little bit about Lie groups.

3.2 Lie Groups

The basic idea of a Lie group is that it is a group which is also a manifold, and that the group structure and the manifold structure are compatible. In this setting, compatibility is going to mean that, appropriately interpreted, the group operations are smooth maps.

Before actually stating the definition, the examples to have in mind are matrix groups: $\text{GL}_n(\mathbb{R}), \text{GL}_n(\mathbb{C}), \text{GL}_n(\mathbb{H}), \text{SL}_n(\mathbb{R}), \text{SL}_n(\mathbb{C}), \text{O}(n), \text{SO}(n), \text{U}(n), \text{SU}(n), \text{Sp}(n)$, the Heisenberg group $\left\{ \begin{bmatrix} 1 & a & c \\ 0 & 1 & b \\ 0 & 0 & 1 \end{bmatrix} : a, b, c \in \mathbb{R} \right\}$, etc.

For example, $\mathrm{GL}_n(\mathbb{C})$ is an open subset of $\mathrm{Mat}_{n \times n}(\mathbb{C}) \cong \mathbb{C}^{n^2} \cong \mathbb{R}^{2n^2}$, so it is certainly a $2n^2$ -dimensional manifold, and the entries of the product of two matrices can be written as (quadratic) polynomials in the entries of the terms, so this is smooth. Moreover, by the adjugate formula $A^{-1} = \frac{1}{\det A} \mathrm{adj}(A)$, the entries of A^{-1} are rational functions of the entries of A with non-vanishing denominators, so taking the inverse of a matrix is also a smooth map.

Moreover, as we saw in [Example 1.3.12](#), the unitary group $\mathrm{U}(n)$ is the level set over a regular value of a smooth map defined on $\mathrm{Mat}_{n \times n}(\mathbb{C})$, so this is also a manifold, and the same argument shows that the group operation and taking inverses are smooth.

Definition 3.2.1. A *Lie group* G is a group which is also a differentiable manifold so that the map $G \times G \rightarrow G$ given by $(g, h) \mapsto gh^{-1}$ is smooth.

The map that shows up in this definition is a slightly funny one; it's maybe more common to require that the maps

- $G \times G \rightarrow G$ given by $(g, h) \mapsto gh$ and
- $G \rightarrow G$ given by $g \mapsto g^{-1}$

are smooth.

Exercise 3.2.2. Show that the above two maps being smooth is equivalent to the single map $(g, h) \mapsto gh^{-1}$ being smooth.

Example 3.2.3. Consider the collection of *affine maps* $\varphi_{a,b} : \mathbb{R} \rightarrow \mathbb{R}$ of the real line, given by $\varphi_{a,b}(t) = at + b$.

The composition of two affine maps is affine:

$$(\varphi_{c,d} \circ \varphi_{a,b})(t) = \varphi_{c,d}(at + b) = c(at + b) + d = act + bc + d. \quad (3.1)$$

Also, if $a \neq 0$ then $\varphi_{a,b}$ is invertible, with inverse $\varphi_{a,b}^{-1} = \varphi_{1/a, -b/a}$. For example,

$$(\varphi_{1/a, -b/a} \circ \varphi_{a,b})(t) = \varphi_{1/a, -b/a}(at + b) = \frac{1}{a}(at + b) - \frac{b}{a} = t + \frac{b}{a} - \frac{b}{a} = t.$$

Notice that, in (3.1), if the two maps being composed are invertible (i.e., $a \neq 0$ and $c \neq 0$) then their composition is also invertible ($ac \neq 0$). Therefore, the collection of invertible affine maps forms a group $\mathrm{Aff}(\mathbb{R})$. To see that it is a Lie group, we just need to check that

$$(\varphi_{a,b}, \varphi_{c,d}) \mapsto \varphi_{a,b} \circ \varphi_{c,d}^{-1} = \varphi_{a,b} \circ \varphi_{1/c, -d/c} = \varphi_{a/c, -ad/c+b}$$

is smooth, which it is since a/c and $-ad/c + b$ are smooth functions of a, b, c, d .

Thinking projectively, we can identify \mathbb{R} with the affine subspace $\left\{ \begin{bmatrix} t \\ 1 \end{bmatrix} : t \in \mathbb{R} \right\} \subset \mathbb{R}^2$ and, since

$$\begin{bmatrix} a & b \\ 0 & 1 \end{bmatrix} \begin{bmatrix} t \\ 1 \end{bmatrix} = \begin{bmatrix} at + b \\ 1 \end{bmatrix},$$

we can represent $\mathrm{Aff}(\mathbb{R})$ as the subgroup of $\mathrm{GL}_2(\mathbb{R})$ given by $\left\{ \begin{bmatrix} a & b \\ 0 & 1 \end{bmatrix} : a \neq 0 \right\}$. We can also see that $\mathrm{Aff}(\mathbb{R})$ is a Lie group from this perspective since it is a closed subgroup of the Lie group $\mathrm{GL}_2(\mathbb{R})$.

Definition 3.2.4. Define *left* and *right translation* by $g \in G$ to be the maps $L_g : G \rightarrow G$ and $R_g : G \rightarrow G$ defined by $L_g(h) = gh$ and $R_g(h) = hg$.

Notice that smoothness of L_g and R_g follows from smoothness of $(g, h) \mapsto gh^{-1}$.

3.3 Lie Algebras

One of the beautiful things about Lie groups is that the algebraic structure of the group and the geometric structure of the underlying manifold are closely related, so we can often learn things about the geometry by studying the algebra, or *vice versa*.

For example, the left multiplication map $L_g : G \rightarrow G$ is a diffeomorphism: we already know it's smooth, and its inverse is clearly $L_{g^{-1}}$, which is also smooth. Moreover, if $e \in G$ is the identity element, then $L_g(e) = g$, so the family of L_g diffeomorphisms will map the identity element of G to any other element of G that we like (equivalently, the action of G on itself by left multiplication is transitive). So even though from the algebraic perspective G has a distinguished point e , from the geometric perspective all points look like all other points (since they all look like the identity).

Importantly, $(dL_g)_e : T_e G \rightarrow T_g G$ is a linear isomorphism, so we can explicitly identify every tangent space with the tangent space to the identity. Note, first of all, that this implies the tangent bundle is trivial: $TG \cong G \times T_e G \cong G \times \mathbb{R}^n$, where n is the dimension of the group. Moreover, it tells us that it might be important to understand the tangent space to the identity.

Indeed, as we will shortly see, the tangent space to the identity has extra algebraic structure, and simply-connected Lie groups are completely determined by this algebraic structure.

The specific algebraic structure is that of a *Lie algebra*, which we've encountered before in [Section 1.6](#) when talking about Lie brackets of vector fields, but now we give the formal definition:

Definition 3.3.1. A real vector space \mathfrak{g} is a *Lie algebra* if it has a bilinear operator $[\cdot, \cdot] : \mathfrak{g} \times \mathfrak{g} \rightarrow \mathfrak{g}$ (the bracket) such that, for all $x, y, z \in \mathfrak{g}$,

- $[x, x] = 0$ (alternativity)
- $[[x, y], z] + [[y, z], x] + [[z, x], y] = 0$ (Jacobi identity).

Example 3.3.2. As we saw in [Example 1.6.7](#), \mathbb{R}^3 forms a Lie algebra with the bracket operation $[v, w] = v \times w$.

Example 3.3.3. As we saw in [Example 1.6.10](#), the space $\mathfrak{so}(3)$ of skew-symmetric 3×3 matrices is a Lie algebra with bracket $[A, B] = AB - BA$.

Example 3.3.4. More generally $\text{Mat}_{n \times n}(\mathbb{R})$ (usually called $\mathfrak{gl}_n(\mathbb{R})$ in this context) is a Lie algebra with the bracket given by the matrix commutator: $[A, B] = AB - BA$. Notice that, by the previous example, $\mathfrak{so}(3)$ is a sub-Lie algebra of $\mathfrak{gl}_n(\mathbb{R})$.

Example 3.3.5. As you showed in HW 1 #4, for any manifold M , $\mathfrak{X}(M)$ is a Lie algebra, where the bracket is the Lie bracket.

By the last example, we know that for any Lie group G the collection of all vector fields $\mathfrak{X}(G)$ is a Lie algebra. However, this is not the Lie algebra we're after when we say *the* Lie algebra of G : first, because Lie

groups aren't special in this regard ($\mathfrak{X}(M)$ is a Lie algebra for any manifold), and second because $\mathfrak{X}(M)$ is a (non-trivial) $C^\infty(M)$ -module, and hence infinite-dimensional as a real vector space.

Building on Examples 1.5.3, 1.6.8, and 1.6.10, we can get a *finite-dimensional* sub-Lie algebra of $\mathfrak{X}(G)$ by focusing on the left-invariant vector fields:

Definition 3.3.6. A vector field X on G (by assumption not necessarily smooth) is *left-invariant* if, for all $g \in G$, $dL_g X = X$. More precisely, this means that for each $h \in G$,

$$(dL_g)_h(X(h)) = X(gh).$$

Theorem 3.3.7. Suppose G is a Lie group and \mathfrak{g} is the set of left-invariant vector fields on G . Then

- (i) \mathfrak{g} is a real vector space and the evaluation map $\text{ev}_e : \mathfrak{g} \rightarrow T_e G$ given by $\text{ev}_e(X) = X(e)$ is a vector space isomorphism. In particular, this implies that $\dim \mathfrak{g} = \dim T_e G = \dim G$.
- (ii) Elements of \mathfrak{g} are smooth vector fields.
- (iii) For $X, Y \in \mathfrak{g}$, $[X, Y] \in \mathfrak{g}$, where $[\cdot, \cdot]$ is the usual Lie bracket of vector fields on manifolds.
- (iv) $(\mathfrak{g}, [\cdot, \cdot])$ is a Lie algebra.

Proof. (i) The map ev_e is certainly linear. It's surjective since, if $v \in T_e G$, we can define a vector field X on G by $X(g) := (dL_g)_e v$. Then X is left-invariant (and hence in \mathfrak{g}) since

$$(dL_g)_h(X(h)) = (dL_g)_h((dL_h)_e v) = (d(L_g \circ L_h))_e(v) = (dL_{gh})_e(v) = X(gh).$$

But then $\text{ev}_e(X) = v$, so ev_e is surjective.

On the other hand, ev_e is also injective: if $\text{ev}_e(X) = \text{ev}_e(Y)$ for $X, Y \in \mathfrak{g}$, then for all $g \in G$ we have

$$X(g) = (dL_g)_e(X(e)) = (dL_g)_e(\text{ev}_e(X)) = (dL_g)_e(\text{ev}_e(Y)) = (dL_g)_e(Y(e)) = Y(g)$$

by left-invariance of X and Y , so we conclude that $\text{ev}_e(X) = \text{ev}_e(Y)$ implies $X = Y$.

(ii) is straightforward but annoying, so we skip it.

(iii) is a consequence of 3.3.8 stated below, and then (iv) is immediate given that the Lie bracket of vector fields is alternating and satisfies the Jacobi identity (which you proved in HW 1 #4). \square

Lemma 3.3.8. Suppose $f : M \rightarrow N$ is smooth and $X_M, Y_M \in \mathfrak{X}(M)$, $X_N, Y_N \in \mathfrak{X}(N)$ so that $df_p X_M(p) = X_N(f(p))$ and $df_p Y_M(p) = Y_N(f(p))$ for all $p \in M$. Then

$$df_p([X_M, Y_M](p)) = [X_N, Y_N](f(p)).$$

In the proof of Theorem 3.3.7 (iii), we apply this lemma with $M = N = G$ and $f = L_g$.

Proof of Lemma 3.3.8. Using the notation $Z_q = Z(q)$ to indicate the tangent vector determined by the vector field Z at a point q , it suffices to show that $(df_p[X_M, Y_M]_p)(\psi) = [X_N, Y_N]_{f(p)}(\psi)$ for any $\psi \in C^\infty(N)$.

This is a computation:

$$\begin{aligned}
(df_p[X_M, Y_M]_p)(\psi) &= [X_M, Y_M]_p(\psi \circ f) = (X_M)_p((Y_M)_p(\psi \circ f)) - (Y_M)_p((X_M)_p(\psi \circ f)) \\
&= (X_M)_p((df_p(Y_M)_p)(\psi)) - (Y_M)_p((df_p(X_M)_p)(\psi)) \\
&= (X_M)_p((Y_N)_{f(p)}(\psi)) - (Y_M)_p((X_N)_{f(p)}(\psi)) \\
&= (X_M)_p(Y_N(\psi) \circ f) - (Y_M)_p(X_N(\psi) \circ f) \\
&= (df_p X_M)(Y_N(\psi)) - (df_p Y_M)(X_N(\psi)) \\
&= (X_N)_{f(p)}(Y_N(\psi)) - (Y_N)_{f(p)}(X_N(\psi)) \\
&= [X_N, Y_N]_{f(p)}(\psi).
\end{aligned}$$

□

Theorem 3.3.7 tells us that any Lie group G has an associated finite-dimensional Lie algebra \mathfrak{g} of left-invariant vector fields; this is usually called *the Lie algebra of G* , typically denoted with a lowercase fraktur version of the name of the group, but sometimes also $\text{Lie}(G)$.

Notice, in particular, that we can also identify \mathfrak{g} with $T_e G$ using the evaluation map, so we could also interpret the Lie bracket as being an operation on this tangent space.

Example 3.3.9. Recall the group $\text{Aff}(\mathbb{R})$ of invertible affine transformations of \mathbb{R} from [Example 3.2.3](#), where we showed that we can represent $\text{Aff}(\mathbb{R})$ as the subgroup $\left\{ \begin{bmatrix} a & b \\ 0 & 1 \end{bmatrix} : a \neq 0 \right\} \subset \text{GL}_2(\mathbb{R})$.

In this representation, $\mathfrak{aff}(\mathbb{R}) \cong T_I \text{Aff}(\mathbb{R})$ has vector space basis $A := \begin{bmatrix} 1 & 0 \\ 0 & 0 \end{bmatrix}$, $B := \begin{bmatrix} 0 & 1 \\ 0 & 0 \end{bmatrix}$; by abuse of notation we can either think of A and B as these specific matrices, or as the left-invariant vector fields on $\text{Aff}(\mathbb{R})$ that they determine.

Exercise 3.3.10. There is an obvious global coordinate chart $\phi: P \rightarrow \text{Aff}(\mathbb{R})$, where $P = \{(u, v) \in \mathbb{R}^2 : u \neq 0\}$ is the plane minus the y -axis, given by $\phi(u, v) = \begin{bmatrix} u & v \\ 0 & 1 \end{bmatrix}$, and associated local coordinate basis $\frac{\partial}{\partial u}, \frac{\partial}{\partial v}$ for each tangent space. The left-invariant vector fields A, B can be written in this basis as

$$A \begin{bmatrix} u & v \\ 0 & 1 \end{bmatrix} = u \frac{\partial}{\partial u} \quad \text{and} \quad B \begin{bmatrix} u & v \\ 0 & 1 \end{bmatrix} = u \frac{\partial}{\partial v}. \quad (3.2)$$

Continuing the example, we will shortly see ([Proposition 3.4.2](#)) that the Lie bracket in the Lie algebra of any matrix group (where we interpret the Lie algebra as the tangent space to the identity, which is just some space of matrices) can be computed as the matrix commutator. Therefore,

$$[A, B] = AB - BA = \begin{bmatrix} 1 & 0 \\ 0 & 0 \end{bmatrix} \begin{bmatrix} 0 & 1 \\ 0 & 0 \end{bmatrix} - \begin{bmatrix} 0 & 1 \\ 0 & 0 \end{bmatrix} \begin{bmatrix} 1 & 0 \\ 0 & 0 \end{bmatrix} = \begin{bmatrix} 0 & 1 \\ 0 & 0 \end{bmatrix} - \begin{bmatrix} 0 & 0 \\ 0 & 0 \end{bmatrix} = B.$$

Anti-commutativity implies $[A, A] = 0$ and $[B, B] = 0$, so $[A, B] = B$ completely determines the Lie algebra structure on $\mathfrak{aff}(\mathbb{R})$. In fact, this turns out to be the *only* non-abelian 2-dimensional Lie algebra.¹

¹That is, the only other 2-dimensional Lie algebra has basis X, Y and trivial bracket $[X, Y] = 0$; for example, this is the Lie algebra of the torus $U(1)^2$, which can be represented by the group of diagonal unitary 2×2 matrices $\left\{ \begin{bmatrix} e^{i\theta_1} & 0 \\ 0 & e^{i\theta_2} \end{bmatrix} \right\} \subset U(2)$.

This Lie algebra is *solvable*, since the derived series terminates:

$$\mathfrak{aff}(\mathbb{R})^{(1)} := [\mathfrak{aff}(\mathbb{R}), \mathfrak{aff}(\mathbb{R})] = \langle B \rangle \quad \text{and} \quad \mathfrak{aff}(\mathbb{R})^{(2)} := [\mathfrak{aff}(\mathbb{R})^{(1)}, \mathfrak{aff}(\mathbb{R})^{(1)}] = [\langle B \rangle, \langle B \rangle] = \{0\}.$$

It is not *nilpotent* since the lower central series continues forever:

$$\mathfrak{aff}(\mathbb{R})_{(1)} := [\mathfrak{aff}(\mathbb{R}), \mathfrak{aff}(\mathbb{R})] = \langle B \rangle, \quad \mathfrak{aff}(\mathbb{R})_{(2)} := [\mathfrak{aff}(\mathbb{R}), \mathfrak{aff}(\mathbb{R})_{(1)}] = [\mathfrak{aff}(\mathbb{R}), \langle B \rangle] = \langle B \rangle, \quad \text{etc.}$$

If α and β are the dual left-invariant 1-forms on $\text{Aff}(\mathbb{R})$, then using [Theorem 2.7.5](#) (Cartan's Magic Formula),

$$\begin{aligned} d\alpha(A, B) &= (\iota_A d\alpha)(B) = (\mathcal{L}_A \alpha)(B) - (d(\iota_A \alpha))(B) = \mathcal{L}_A(\alpha(B)) - \alpha(\mathcal{L}_A B) - (d(1))(B) \\ &= \mathcal{L}_A(0) - \alpha([A, B]) - 0 = -\alpha(B) = 0 \end{aligned}$$

and

$$\begin{aligned} d\beta(A, B) &= (\iota_A d\beta)(B) = (\mathcal{L}_A \beta)(B) - (d(\iota_A \beta))(B) = \mathcal{L}_A(\beta(B)) - \beta(\mathcal{L}_A B) - (d(0))(B) \\ &= \mathcal{L}_A(1) - \beta([A, B]) - 0 = -\beta(B) = 1, \end{aligned}$$

so $d\alpha = 0$ and $d\beta = -\alpha \wedge \beta$.

Just by looking at the coordinate expressions (3.2) for A and B , we could pretty easily have written down the formulas

$$\alpha \begin{bmatrix} u & v \\ 0 & 1 \end{bmatrix} = \frac{1}{u} du \quad \text{and} \quad \beta \begin{bmatrix} u & v \\ 0 & 1 \end{bmatrix} = \frac{1}{u} dv,$$

from which we could have directly computed

$$\begin{aligned} d\alpha &= d\left(\frac{1}{u} du\right) = 0 \\ d\beta &= d\left(\frac{1}{u} dv\right) = -\frac{1}{u^2} du \wedge dv = -\left(\frac{1}{u} du\right) \wedge \left(\frac{1}{u} dv\right) = -\alpha \wedge \beta. \end{aligned}$$

From this perspective, it's clear that $\alpha = d(\log |u|)$.²

3.4 Matrix Groups

Definition 3.4.1. A *matrix group* is a closed subgroup of $\text{GL}_n(\mathbb{R})$ for some n .

You might worry that this excludes groups consisting of complex or quaternionic matrices, but all such groups can be represented as subgroups of some real general linear group. Specifically, define the map $\rho_1: \mathbb{C} \rightarrow \text{Mat}_{2 \times 2}(\mathbb{R})$ by

$$\rho_1(x + yi) = \begin{bmatrix} x & -y \\ y & x \end{bmatrix}.$$

²Because the base-10 logarithm is pretty useless and I'm not a computer scientist, I use \log to denote the natural logarithm (a.k.a. \ln).

This is the right definition to turn multiplication of complex numbers into linear transformations of \mathbb{R} . After all, if $z = x + iy$, then for $w = u + iv$ we have

$$zw = (x + yi)(u + vi) = xu - yv + i(xv + yu)$$

and

$$\rho_1(z) \begin{bmatrix} u \\ v \end{bmatrix} = \begin{bmatrix} x & -y \\ y & x \end{bmatrix} \begin{bmatrix} u \\ v \end{bmatrix} = \begin{bmatrix} xu - yv \\ xv + yu \end{bmatrix}.$$

In turn, we can get a map $\rho_n: \text{Mat}_{n \times n}(\mathbb{C}) \rightarrow \text{Mat}_{2n \times 2n}(\mathbb{R})$ by just applying ρ_1 to each entry, resulting in a real matrix built out of 2×2 blocks of the form $\begin{bmatrix} x & -y \\ y & x \end{bmatrix}$. For example

$$\rho_3 \left(\begin{bmatrix} x_{11} + y_{11}i & x_{12} + y_{12}i & x_{13} + y_{13}i \\ x_{21} + y_{21}i & x_{22} + y_{22}i & x_{23} + y_{23}i \\ x_{31} + y_{31}i & x_{32} + y_{32}i & x_{33} + y_{33}i \end{bmatrix} \right) = \begin{bmatrix} x_{11} & -y_{11} & x_{12} & -y_{12} & x_{13} & -y_{13} \\ y_{11} & x_{11} & y_{12} & x_{12} & y_{13} & x_{13} \\ x_{21} & -y_{21} & x_{22} & -y_{22} & x_{23} & -y_{23} \\ y_{21} & x_{21} & y_{22} & x_{22} & y_{23} & x_{23} \\ x_{31} & -y_{31} & x_{32} & -y_{32} & x_{33} & -y_{33} \\ y_{31} & x_{31} & y_{32} & x_{32} & y_{33} & x_{33} \end{bmatrix}.$$

It's not hard to show that ρ_n preserves invertibility, so $\rho_n(\text{GL}_n(\mathbb{C})) \subset \text{GL}_{2n}(\mathbb{R})$, showing that $\text{GL}_n(\mathbb{C})$ and all of its closed subgroups are matrix groups.

Similarly, we can define $\psi_1: \mathbb{H} \rightarrow \text{Mat}_{2 \times 2}(\mathbb{C})$ by

$$\psi_1(z + wj) = \begin{bmatrix} z & w \\ -\bar{w} & \bar{z} \end{bmatrix}$$

or, written out more completely,

$$\psi_1(a + bi + cj + dk) = \begin{bmatrix} a + bi & c + di \\ -c + di & a - bi \end{bmatrix}.$$

And then we can extend to a map $\psi_n: \text{Mat}_{n \times n}(\mathbb{H}) \rightarrow \text{Mat}_{2n \times 2n}(\mathbb{C})$ as you'd expect. For example,

$$\begin{aligned} \psi_2 \left(\begin{bmatrix} a_{11} + b_{11}i + c_{11}j + d_{11}k & a_{12} + b_{12}i + c_{12}j + d_{12}k \\ a_{21} + b_{21}i + c_{21}j + d_{21}k & a_{22} + b_{22}i + c_{22}j + d_{22}k \end{bmatrix} \right) \\ = \begin{bmatrix} a_{11} + ib_{11} & c_{11} + id_{11} & a_{12} + ib_{12} & c_{12} + id_{12} \\ -c_{11} + id_{11} & a_{11} - ib_{11} & -c_{12} + id_{12} & a_{12} - ib_{12} \\ a_{21} + ib_{21} & c_{21} + id_{21} & a_{22} + ib_{22} & c_{22} + id_{22} \\ -c_{21} + id_{21} & a_{21} - ib_{21} & -c_{22} + id_{22} & a_{22} - ib_{22} \end{bmatrix} \end{aligned}$$

Again, $\psi_n(\text{GL}_n(\mathbb{H})) \subset \text{GL}_{2n}(\mathbb{C})$, and of course by composing ψ_n with ρ_{2n} we can represent any quaternionic matrix group as a subgroup of $\text{GL}_{4n}(\mathbb{R})$.

The nice thing about matrix groups is that the bracket operations on their Lie algebras are particularly simple.

Proposition 3.4.2. *Let $G \leq \text{GL}_n(\mathbb{R})$ be a matrix group, so that $T_I G \subset T_I \text{GL}_n(\mathbb{R}) \cong \text{Mat}_{n \times n}(\mathbb{R})$. Then the bracket operation on \mathfrak{g} , interpreted as an operation on $T_I G$, is just the matrix commutator: for $X, Y \in T_I G \cong \mathfrak{g}$, the bracket $[X, Y] = XY - YX$ where XY just means the product of matrices.*

Proof. We prove this for $G = \mathrm{GL}_n(\mathbb{R})$ and then it follows for subgroups.

We're going to prove this by writing out the matrix multiplication in terms of coordinates; i.e., if $A, B \in \mathrm{Mat}_{n \times n}(\mathbb{R})$, then

$$(AB)_{ij} = \sum_{k=1}^n A_{ik}B_{kj}.$$

But now, if we're thinking of our matrices as tangent vectors at the identity, then they're really differential operators, so we need a way of interpreting these coordinates in that language.

To that end, by analogy with what we did in [Section 2.4.1](#), define $x_{ij} : \mathrm{Mat}_{n \times n}(\mathbb{R}) \rightarrow \mathbb{R}$ by $x_{ij}(A) = A_{ij}$, the (i, j) entry of A . Then if $V \in T_I \mathrm{GL}_n(\mathbb{R}) \cong \mathrm{Mat}_{n \times n}(\mathbb{R})$, we can write $V = \alpha'(0)$ where $\alpha(t) = I + tV$, which stays in $\mathrm{GL}_n(\mathbb{R})$ for small enough t . Therefore, using [Definition 1.2.1](#)

$$V(x_{ij}) = (x_{ij} \circ \alpha)'(0) = \frac{d}{dt} \Big|_{t=0} x_{ij}(\alpha(t)) = \frac{d}{dt} \Big|_{t=0} \alpha(t)_{ij} = \alpha'(0)_{ij} = V_{ij}. \quad (3.3)$$

So this gives a way of translating back and forth between elements of $T_I \mathrm{GL}_n(\mathbb{R})$ as differential operators and as arrays of numbers.

Next, recall the evaluation map $\mathrm{ev}_I : \mathfrak{gl}_n(\mathbb{R}) \rightarrow T_I \mathrm{GL}_n(\mathbb{R})$ given by $\mathrm{ev}_I(X) = X(I)$, the tangent vector at I determined by the left-invariant vector field X . In what follows, to avoid a surplus of parentheses, we will sometimes write $X(I)$ as X_I .

We saw in the proof of [Theorem 3.3.7](#) that ev_I is a vector space isomorphism, so all that remains to do is to show that it is a Lie algebra homomorphism; that is, for any left-invariant vector fields X, Y on $\mathrm{GL}_n(\mathbb{R})$ (i.e., elements of $\mathfrak{gl}_n(\mathbb{R})$), we need to show that

$$\mathrm{ev}_I([X, Y]) = \mathrm{ev}_I(X) \mathrm{ev}_I(Y) - \mathrm{ev}_I(Y) \mathrm{ev}_I(X) = X(I)Y(I) - Y(I)X(I),$$

where on the right hand side we're just multiplying matrices.

Notice, first of all, that for any $g, h \in \mathrm{GL}_n(\mathbb{R})$,

$$(x_{ij} \circ L_g)(h) = x_{ij}(L_g(h)) = x_{ij}(gh) = (gh)_{ij} = \sum_{k=1}^n g_{ik}h_{kj} = \sum_{k=1}^n x_{ik}(g)x_{kj}(h);$$

in other words,

$$x_{ij} \circ L_g = \sum_{k=1}^n x_{ik}(g)x_{kj}. \quad (3.4)$$

We're going to look now at the function $Y(x_{ij})$, which we can specify by saying how it evaluates at each $g \in \mathrm{GL}_n(\mathbb{R})$. By left-invariance of Y together with (3.3) and (3.4),

$$(Y(x_{ij})) (g) = Y_g(x_{ij}) = ((dL_g)_I Y_I)(x_{ij}) = Y_I(x_{ij} \circ L_g) = \sum_{k=1}^n x_{ik}(g)Y_I(x_{kj}) = \sum_{k=1}^n x_{ik}(g)Y(I)_{kj}.$$

This (and the corresponding statement for X) gives us the necessary tool to determine the (i, j) component of $\mathrm{ev}_I([X, Y])$:

$$\begin{aligned} (\mathrm{ev}_I([X, Y]))_{ij} &= [X, Y]_I(x_{ij}) = X_I(Y(x_{ij})) - Y_I(X(x_{ij})) \\ &= X_I \left(\sum_{k=1}^n x_{ik}Y(I)_{kj} \right) - Y_I \left(\sum_{k=1}^n x_{ik}X(I)_{kj} \right) = \sum_{k=1}^n (X_I(x_{ik}Y(I)_{kj}) - Y_I(x_{ik}X(I)_{kj})) \end{aligned}$$

by linearity of the operators X_I and Y_I . Of course, $Y(I)_{kj}$ and $X(I)_{kj}$ are constant, so the Leibniz rule gives us

$$\sum_{k=1}^n (X_I(x_{ik})Y(I)_{kj} - Y_I(x_{ik})X(I)_{kj}) = \sum_{k=1}^n (X(I)_{ik}Y(I)_{kj} - Y(I)_{ik}X(I)_{kj}) = (X(I)Y(I) - Y(I)X(I))_{ij},$$

where, as a reminder, I'm freely switching between the notations $X(I)$ and X_I for the tangent vector at I determined by the vector field X . So this shows that the left and right sides of the equation in the statement of the theorem agree componentwise. \square

This now justifies the computational simplifications we did in looking at the Lie algebras of $U(d)$ (Example 1.6.8), $SO(d)$ (Example 1.6.10), and $Aff(\mathbb{R})$ (Example 3.3.9).

Proposition 3.4.3. $\mathfrak{so}(3) \cong \mathfrak{su}(2) \cong \mathfrak{sp}(1)$.

Proof. In each case this is the Lie algebra of a matrix group, so Proposition 3.4.2 guarantees it is isomorphic to the tangent space at the identity with the bracket given by the matrix commutator. This is the interpretation we will use to prove the proposition.

For example, $T_I SO(3)$ consists of the 3×3 skew-symmetric matrices, which we saw in Example 1.6.10 is isomorphic to (\mathbb{R}^3, \times) .

We've seen in Example 1.5.3 that $T_I U(d)$ consists of the skew-Hermitian $d \times d$ matrices; I claim without proof that the determinant-1 condition translates to a trace-0 condition, so $T_I SU(d)$ is the space of traceless skew-Hermitian matrices.

Finally, thinking of $Sp(1) \cong S^3$ as the 1×1 quaternionic matrices which preserve the symplectic inner product (i.e., the quaternionic analog of $O(d)$ or $U(d)$), the same sort of argument that gave the tangent spaces at the identity of $O(d)$ and $U(d)$ will imply that $T_I Sp(1)$ consists of quaternionic skew-Hermitian 1×1 matrices; in other words, the single entry must be a purely imaginary quaternion $ai + bj + ck$.

We can fairly easily write down bases for $\mathfrak{so}(3)$, $\mathfrak{su}(2)$, and $\mathfrak{sp}(1)$:

$$\begin{aligned}\mathfrak{so}(3) &= \text{span} \left\{ \begin{bmatrix} 0 & 0 & 0 \\ 0 & 0 & -1 \\ 0 & 1 & 0 \end{bmatrix}, \begin{bmatrix} 0 & 0 & 1 \\ 0 & 0 & 0 \\ -1 & 0 & 0 \end{bmatrix}, \begin{bmatrix} 0 & -1 & 0 \\ 1 & 0 & 0 \\ 0 & 0 & 0 \end{bmatrix} \right\} \\ \mathfrak{su}(2) &= \text{span} \left\{ \frac{1}{2} \begin{bmatrix} i & 0 \\ 0 & -i \end{bmatrix}, \frac{1}{2} \begin{bmatrix} 0 & 1 \\ -1 & 0 \end{bmatrix}, \frac{1}{2} \begin{bmatrix} 0 & i \\ i & 0 \end{bmatrix} \right\} \\ \mathfrak{sp}(1) &= \text{span} \left\{ \frac{i}{2}, \frac{j}{2}, \frac{k}{2} \right\}.\end{aligned}$$

If we use $\{A_{1,g}, A_{2,g}, A_{3,g}\}$ to denote the basis for $g \in \{\mathfrak{so}(3), \mathfrak{su}(2), \mathfrak{sp}(1)\}$, then it is easy to check that

$$[A_{1,g}, A_{2,g}] = A_{3,g}, \quad [A_{2,g}, A_{3,g}] = A_{1,g}, \quad [A_{3,g}, A_{1,g}] = A_{2,g}.$$

For example,

$$[A_{1,\mathfrak{so}(3)}, A_{2,\mathfrak{so}(3)}] = \begin{bmatrix} 0 & 0 & 0 \\ 0 & 0 & -1 \\ 0 & 1 & 0 \end{bmatrix} \begin{bmatrix} 0 & 0 & 1 \\ 0 & 0 & 0 \\ -1 & 0 & 0 \end{bmatrix} - \begin{bmatrix} 0 & 0 & 1 \\ 0 & 0 & 0 \\ -1 & 0 & 0 \end{bmatrix} \begin{bmatrix} 0 & 0 & 0 \\ 0 & 0 & -1 \\ 0 & 1 & 0 \end{bmatrix} = \begin{bmatrix} 0 & -1 & 0 \\ 1 & 0 & 0 \\ 0 & 0 & 0 \end{bmatrix} = A_{3,\mathfrak{so}(3)}$$

and

$$[A_{3,\mathfrak{sp}(1)}, A_{1,\mathfrak{sp}(1)}] = \frac{k}{2} \frac{i}{2} - \frac{i}{2} \frac{k}{2} = \frac{j}{4} + \frac{j}{4} = \frac{j}{2} = A_{2,\mathfrak{sp}(1)}.$$

Therefore, for $\{\mathfrak{g}, \mathfrak{g}'\} \subset \{\mathfrak{so}(3), \mathfrak{su}(2), \mathfrak{sp}(1)\}$, linearly extending the identifications $A_{i,\mathfrak{g}} \mapsto A_{i,\mathfrak{g}'}$ defines a Lie algebra isomorphism. \square

[Proposition 3.4.3](#) is an example of what is sometimes called an *accidental isomorphism*, which in this case boil down to the fact that there just aren't that many 3-dimensional Lie algebras. In general $\mathfrak{su}(n+1)$, $\mathfrak{so}(2n+1)$, and $\mathfrak{sp}(n)$ belong to three different infinite families of Lie algebras: they are A_n , B_n , and C_n , respectively. In this language, [Proposition 3.4.3](#) says that $A_1 \cong B_1 \cong C_1$.

3.5 Adjoint and Coadjoint Representations

For each $g \in G$, let $C_g: G \rightarrow G$ denote conjugation by g , namely $C_g(h) = ghg^{-1}$. Notice, in particular, that $C_g(e) = geg^{-1} = e$ for all $g \in G$, so these maps are all diffeomorphisms fixing the identity, and hence $(dC_g)_e: T_e G \rightarrow T_e G$ for all $g \in G$.

When G is a matrix group it's easy to get a formula for $(dC_g)_e$: since $C_g = L_g \circ R_{g^{-1}}$, we know that $dC_g = d(L_g \circ R_{g^{-1}}) = dL_g \circ dR_{g^{-1}}$, so it suffices to find nice formulas for dL_g and $dR_{g^{-1}}$. For example, if $v \in T_h G$ and $v = \alpha'(0)$ for some smooth curve $\alpha(t)$ with $\alpha(0) = h$, then

$$(dL_g)_h v = (L_g \circ \alpha)'(0) = \left. \frac{d}{dt} \right|_{t=0} L_g(\alpha(t)) = \left. \frac{d}{dt} \right|_{t=0} g\alpha(t) = g\alpha'(0) = gv,$$

where the operation here is matrix multiplication, which makes sense when we interpret $T_h G \subset T_h \mathrm{GL}_n(\mathbb{R}) = \mathrm{Mat}_{n \times n}(\mathbb{R})$.

The analogous statement is true for $dR_{g^{-1}}$, so we see that, on matrix groups, $(dC_g)_e v = gv g^{-1}$; that is, this is just the conjugation action of the group on its tangent space at the identity.

Of course, we can identify $T_e G$ with the Lie algebra \mathfrak{g} , so we can also think of $(dC_g)_e$ as a map $\mathfrak{g} \rightarrow \mathfrak{g}$, but in this guise we usually call the map Ad_g rather than $(dC_g)_e$; that is, we have $\mathrm{Ad}_g: \mathfrak{g} \rightarrow \mathfrak{g}$, called the *adjoint action* of $g \in G$ on \mathfrak{g} . Said yet another way, the following is well-defined:

Definition 3.5.1. Let G be a Lie group with Lie algebra \mathfrak{g} . Let $\mathrm{Aut}(\mathfrak{g})$ be the group of invertible linear transformations of (the vector space) \mathfrak{g} (sometimes instead called $\mathrm{GL}(\mathfrak{g})$). The map $\mathrm{Ad}: G \rightarrow \mathrm{Aut}(\mathfrak{g})$ defined by $g \mapsto \mathrm{Ad}_g$ is called the *adjoint representation* of G .

Notice that Ad is a Lie group homomorphism: since

$$C_{gh}(k) = (gh)k(gh)^{-1} = ghkh^{-1}g^{-1} = C_g(C_h(k)) = (C_g \circ C_h)(k),$$

we see that $C_{gh} = C_g \circ C_h$, so

$$\mathrm{Ad}_{gh} = (dC_{gh})_e = (d(C_g \circ C_h))_e = (dC_g)_{C_h(e)} (dC_h)_e = (dC_g)_e (dC_h) = \mathrm{Ad}_g \mathrm{Ad}_h.$$

One of the key virtues of the adjoint representation is that it defines a homomorphism from *any* Lie group to a matrix group. Moreover, $\ker \mathrm{Ad} = Z(G)$, the center of G , so the first isomorphism theorem implies that $\mathrm{Ad}(G) \cong G/Z(G)$; in particular, this shows that any centerless Lie group can be realized as a matrix group.

Example 3.5.2. Recall again $\text{Aff}(\mathbb{R})$ from Example 3.2.3 and its Lie algebra $\mathfrak{aff}(\mathbb{R})$, which we saw in Example 3.3.9 has basis

$$A = \begin{bmatrix} 1 & 0 \\ 0 & 0 \end{bmatrix}, \quad B = \begin{bmatrix} 0 & 1 \\ 0 & 0 \end{bmatrix}$$

and the Lie bracket satisfies $[A, B] = B$.

Now, suppose $\begin{bmatrix} u & v \\ 0 & 1 \end{bmatrix} \in \text{Aff}(\mathbb{R})$. Then we can specify the adjoint action of $\begin{bmatrix} u & v \\ 0 & 1 \end{bmatrix}$ on $\mathfrak{aff}(\mathbb{R})$ by determining what it does to A and B :

$$\begin{aligned} \text{Ad}_{\begin{bmatrix} u & v \\ 0 & 1 \end{bmatrix}}(A) &= \begin{bmatrix} u & v \\ 0 & 1 \end{bmatrix} \begin{bmatrix} 1 & 0 \\ 0 & 0 \end{bmatrix} \begin{bmatrix} u & v \\ 0 & 1 \end{bmatrix}^{-1} = \begin{bmatrix} u & v \\ 0 & 1 \end{bmatrix} \begin{bmatrix} 1 & 0 \\ 0 & 0 \end{bmatrix} \begin{bmatrix} \frac{1}{u} & -\frac{v}{u} \\ 0 & 1 \end{bmatrix} = \begin{bmatrix} 1 & -v \\ 0 & 0 \end{bmatrix} = A - vB \\ \text{Ad}_{\begin{bmatrix} u & v \\ 0 & 1 \end{bmatrix}}(B) &= \begin{bmatrix} u & v \\ 0 & 1 \end{bmatrix} \begin{bmatrix} 0 & 1 \\ 0 & 0 \end{bmatrix} \begin{bmatrix} u & v \\ 0 & 1 \end{bmatrix}^{-1} = \begin{bmatrix} u & v \\ 0 & 1 \end{bmatrix} \begin{bmatrix} 0 & 1 \\ 0 & 0 \end{bmatrix} \begin{bmatrix} \frac{1}{u} & -\frac{v}{u} \\ 0 & 1 \end{bmatrix} = \begin{bmatrix} 0 & u \\ 0 & 0 \end{bmatrix} = uB. \end{aligned}$$

In other words, the matrix for $\text{Ad}_{\begin{bmatrix} u & v \\ 0 & 1 \end{bmatrix}}$ with respect to the (ordered) basis $\{A, B\}$ is $\begin{bmatrix} 1 & 0 \\ -v & u \end{bmatrix}$, with eigenvalues 1 and u and corresponding eigenvectors $\begin{bmatrix} u-1 \\ v \end{bmatrix} = (u-1)A + vB$ and $\begin{bmatrix} 0 \\ 1 \end{bmatrix} = B$.

In various circumstances, some of which we may encounter later, it is more convenient to dualize, meaning we want to look at the induced action on \mathfrak{g}^* .

Given $g \in G$, the map $\text{Ad}_g : \mathfrak{g} \rightarrow \mathfrak{g}$ has, in the spirit of Section 2.9.1, an associated dual map $(\text{Ad}_g)^* : \mathfrak{g}^* \rightarrow \mathfrak{g}^*$ defined by

$$(\text{Ad}_g)^*(\xi)(X) = \xi(\text{Ad}_g(X))$$

for any $\xi \in \mathfrak{g}^*$ and any $X \in \mathfrak{g}$, but this does not define a homomorphism $G \rightarrow \text{Aut}(\mathfrak{g}^*)$: for any $g, h \in G$ and $\xi \in \mathfrak{g}^*$,

$$(\text{Ad}_{gh})^*(\xi)(X) = \xi(\text{Ad}_{gh}(X)) = \xi(\text{Ad}_g(\text{Ad}_h(X))) = (\text{Ad}_g)^*(\xi)(\text{Ad}_h(X)) = (\text{Ad}_h)^*((\text{Ad}_g)^*(\xi))(X)$$

for any $X \in \mathfrak{g}$. In other words, $(\text{Ad}_{gh})^* = (\text{Ad}_h)^* \circ (\text{Ad}_g)^*$, which is not generally equal to $(\text{Ad}_g)^* \circ (\text{Ad}_h)^*$.

To fix this and get an honest Lie group homomorphism, we need to twist by the involution $g \mapsto g^{-1}$ (in fancy categorical language, I believe this is the canonical isomorphism of G with G^{op} and the dual map really defines a homomorphism $G^{\text{op}} \rightarrow \text{Aut}(\mathfrak{g}^*)$):

Definition 3.5.3. Let G be a Lie group, \mathfrak{g} its Lie algebra, and $g \in G$. The *coadjoint action* of g on \mathfrak{g}^* is the map $\text{Ad}_g^* : \mathfrak{g}^* \rightarrow \mathfrak{g}^*$ defined on any $\xi \in \mathfrak{g}^*$ by

$$\text{Ad}_g^*(\xi)(X) := \xi(\text{Ad}_{g^{-1}}(X))$$

for all $X \in \mathfrak{g}$. (In other words, $\text{Ad}_g^* = (\text{Ad}_{g^{-1}})^*$, where the position of the parentheses is quite important.)

In turn, this defines the *coadjoint representation* $\text{Ad}^* : G \rightarrow \text{Aut}(\mathfrak{g}^*)$ given by $g \mapsto \text{Ad}_g^*$.

One virtue of the coadjoint action is that each orbit (called a *coadjoint orbit*) has a natural symplectic structure.

Example 3.5.4 (Example 3.5.2 continued). The matrix of the coadjoint action

$$\text{Ad}_{\begin{bmatrix} u & v \\ 0 & 1 \end{bmatrix}}^* = \left(\text{Ad}_{\begin{bmatrix} u & v \\ 0 & 1 \end{bmatrix}} \right)^*: \mathfrak{aff}(\mathbb{R})^* \rightarrow \mathfrak{aff}(\mathbb{R})^*$$

with respect to the dual basis $\{\alpha, \beta\}$ from Example 3.3.9 is $\begin{bmatrix} 1 & 0 \\ \frac{v}{u} & \frac{1}{u} \end{bmatrix}^T = \begin{bmatrix} 1 & \frac{v}{u} \\ 0 & \frac{1}{u} \end{bmatrix}$ since $\begin{bmatrix} u & v \\ 0 & 1 \end{bmatrix}^{-1} = \begin{bmatrix} \frac{1}{u} & -\frac{v}{u} \\ 0 & 1 \end{bmatrix}$; equivalently,

$$\text{Ad}_{\begin{bmatrix} u & v \\ 0 & 1 \end{bmatrix}}^*(\alpha) = \alpha \quad \text{and} \quad \text{Ad}_{\begin{bmatrix} u & v \\ 0 & 1 \end{bmatrix}}^*(\beta) = \frac{v}{u}\alpha + \frac{1}{u}\beta.$$

The eigenvalues of $\text{Ad}_{\begin{bmatrix} u & v \\ 0 & 1 \end{bmatrix}}^*$ are 1 and $\frac{1}{u}$ with corresponding eigenvectors $\begin{bmatrix} 1 \\ 0 \end{bmatrix} = \alpha$ and $\begin{bmatrix} v \\ 1-u \end{bmatrix} = v\alpha + (1-u)\beta$.

Example 3.5.5. Let's understand the adjoint representation of $\text{SO}(3)$. Recall from Example 1.6.10 that $\mathfrak{so}(3)$ consists of skew-symmetric 3×3 matrices and is isomorphic as a Lie algebra to (\mathbb{R}^3, \times) via the map

$$\phi: (x, y, z) \mapsto \begin{bmatrix} 0 & -z & y \\ z & 0 & -x \\ -y & x & 0 \end{bmatrix}.$$

Under this identification of $\mathfrak{so}(3)$ with \mathbb{R}^3 , the adjoint representation should give an action of $\text{SO}(3)$ on \mathbb{R}^3 . Perhaps surprisingly (and, then again, perhaps not), this action turns out to be the usual action of $\text{SO}(3)$ by matrix-vector multiplication. More precisely, for $A \in \text{SO}(3)$ and $v \in \mathbb{R}^3$,

$$\text{Ad}_A(\phi(v)) = \phi(Av).$$

Rather than give a full proof, let's see that it works for a simple subgroup of $\text{SO}(3)$. Specifically, let

$A(t) = \begin{bmatrix} \cos t & -\sin t & 0 \\ \sin t & \cos t & 0 \\ 0 & 0 & 1 \end{bmatrix}$ and consider the subgroup $H = \{A(t) : t \in \mathbb{R}\}$, which acts by rotation around the

z -axis. Then, for $v = \begin{bmatrix} x \\ y \\ z \end{bmatrix}$, we have

$$\begin{aligned} \text{Ad}_{A(t)}(\phi(v)) &= A(t)\phi(v)A(t)^{-1} = \begin{bmatrix} \cos t & -\sin t & 0 \\ \sin t & \cos t & 0 \\ 0 & 0 & 1 \end{bmatrix} \begin{bmatrix} 0 & -z & y \\ z & 0 & -x \\ -y & x & 0 \end{bmatrix} \begin{bmatrix} \cos t & \sin t & 0 \\ -\sin t & \cos t & 0 \\ 0 & 0 & 1 \end{bmatrix} \\ &= \begin{bmatrix} 0 & -z & x \sin t + y \cos t \\ z & 0 & -x \cos t + y \sin t \\ -x \sin t - y \cos t & x \cos t - y \sin t & 0 \end{bmatrix} = \phi\left(\begin{bmatrix} x \cos t - y \sin t \\ x \sin t + y \cos t \\ z \end{bmatrix}\right) = \phi(A(t)v). \end{aligned}$$

So we see that, at least when restricted to H , the adjoint representation of $\text{SO}(3)$ agrees with the usual rotation representation.

3.6 Lie Algebra Homomorphisms

We've just seen that a Lie group always acts on its Lie algebra by the adjoint action. Now we start to develop another connection between Lie groups and Lie algebras, one already hinted at:

Theorem 3.6.1. *Let G and H be Lie groups with Lie algebras \mathfrak{g} and \mathfrak{h} , respectively, and with G simply-connected. Let $\psi : \mathfrak{g} \rightarrow \mathfrak{h}$ be a Lie algebra homomorphism. Then there exists a unique Lie group homomorphism $\phi : G \rightarrow H$ so that $d\phi = \psi$.*

One way to interpret this is that Lie group homomorphisms are (essentially) uniquely determined by their behavior in arbitrarily small neighborhoods of the identity (since they're determined by their linearizations at the identity). A more or less immediate corollary of the theorem is that simply-connected Lie groups are uniquely determined by their Lie algebras:

Corollary 3.6.2. *If G and H are simply-connected and have isomorphic Lie algebras, then $G \cong H$.*

A first step in the proof of [Theorem 3.6.1](#) is to show that Lie group homomorphisms induce Lie algebra homomorphisms:

Proposition 3.6.3. *Let G and H be Lie groups, and let $\phi : G \rightarrow H$ be a Lie group homomorphism. Then $d\phi : \mathfrak{g} \rightarrow \mathfrak{h}$ is a Lie algebra homomorphism.*

Before proving this, it's probably a good idea to pause for a moment and decipher notation. By definition, \mathfrak{g} is the collection of left-invariant vector fields on G , so if X is a left-invariant vector field on G , then $d\phi(X)$ is the vector field on H given by $(d\phi)_g X(g)$ at each point $\phi(g)$ in the image of ϕ , and then extended to a left-invariant vector field on all of H .

Exercise 3.6.4. Prove that the map $d\phi : \mathfrak{g} \rightarrow \mathfrak{h}$ described in the previous paragraph is well-defined; that is, if $\phi(g_1) = \phi(g_2)$, then $(d\phi)_{g_1} X(g_1) = (d\phi)_{g_2} X(g_2)$.

Proof of Proposition 3.6.3. First, we want to show that $d\phi(X)$ is left-invariant on the image of ϕ , and hence extends to an element of \mathfrak{h} . To that end, if $X \in \mathfrak{g}$, then for each $g_1, g_2 \in G$,

$$\begin{aligned} dL_{\phi(g_1)} d\phi_{g_2} X(g_2) &= d(L_{\phi(g_1)} \circ \phi)_{g_2} X(g_2) = d(\phi \circ L_{g_1})_{g_2} X(g_2) \\ &= (d\phi)_{L_{g_1}(g_2)} (dL_{g_1})_{g_2} X(g_2) = (d\phi)_{g_1 g_2} X(g_1 g_2), \end{aligned} \quad (3.5)$$

where we used the Chain Rule in the first and third equalities and the left-invariance of X in the last equality. The second equality follows because ϕ is a Lie group homomorphism in the second; more explicitly, for any $g_1, g_2 \in G$,

$$(L_{\phi(g_1)} \circ \phi)(g_2) = L_{\phi(g_1)}(\phi(g_2)) = \phi(g_1)\phi(g_2) = \phi(g_1 g_2) = \phi(L_{g_1}(g_2)) = (\phi \circ L_{g_1})(g_2),$$

so we see that $L_{\phi(g_1)} \circ \phi = \phi \circ L_{g_1}$ and the second equality in (3.5) follows.

Thus, we have shown that $d\phi$ really maps \mathfrak{g} to \mathfrak{h} . To see that it is an Lie algebra homomorphism, recall from [Lemma 3.3.8](#) that, for $X, Y \in \mathfrak{g}$ and $g \in G$,

$$d\phi_g([X, Y](g)) = [d\phi_g X, d\phi_g Y](\phi(g)),$$

or, more poetically, $d\phi[X, Y] = [d\phi X, d\phi Y]$, which is what it means for $d\phi$ to be a Lie algebra homomorphism. \square

Example 3.6.5. Consider the inclusion $i: \mathrm{SO}(n) \rightarrow \mathrm{SO}(n+1)$ given by

$$i(A) = \begin{bmatrix} A & \mathbf{0} \\ \mathbf{0}^T & 1 \end{bmatrix},$$

where the matrix on the right is an $(n+1) \times (n+1)$ block matrix, where $\mathbf{0}$ denotes the $n \times 1$ zero matrix.

From [Example 1.6.10](#) we know that we can interpret elements of $\mathfrak{so}(n)$ as $n \times n$ skew-symmetric matrices. If X is such a matrix so that $X = \alpha'(0)$ with $\alpha(0) = I_{d \times d}$, then

$$di(X) = (di)_{I_{d \times d}}(X) = (i \circ \alpha)'(0) = \frac{d}{dt} \Big|_{t=0} i(\alpha(t)) = \frac{d}{dt} \Big|_{t=0} \begin{bmatrix} \alpha(t) & \mathbf{0} \\ \mathbf{0}^T & 1 \end{bmatrix} = \begin{bmatrix} \alpha'(0) & \mathbf{0} \\ \mathbf{0}^T & 0 \end{bmatrix} = \begin{bmatrix} X & \mathbf{0} \\ \mathbf{0}^T & 0 \end{bmatrix},$$

which is indeed a skew-symmetric $(n+1) \times (n+1)$ matrix, and so can be interpreted as an element of $\mathfrak{so}(n+1)$. Moreover, for $X, Y \in \mathfrak{so}(n)$,

$$di([X, Y]) = \begin{bmatrix} [X, Y] & \mathbf{0} \\ \mathbf{0}^T & 0 \end{bmatrix} = \begin{bmatrix} X & \mathbf{0} \\ \mathbf{0}^T & 0 \end{bmatrix} \begin{bmatrix} Y & \mathbf{0} \\ \mathbf{0}^T & 0 \end{bmatrix} - \begin{bmatrix} Y & \mathbf{0} \\ \mathbf{0}^T & 0 \end{bmatrix} \begin{bmatrix} X & \mathbf{0} \\ \mathbf{0}^T & 0 \end{bmatrix} = [di(X), di(Y)],$$

so this really is a Lie algebra homomorphism.

Example 3.6.6. Consider the coordinate chart $f: \mathrm{Aff}(\mathbb{R}) \rightarrow \mathrm{GL}_2(\mathbb{R})$ given by $f(\varphi_{a,b}) := \begin{bmatrix} a & b \\ 0 & 1 \end{bmatrix}$ implicitly defined in [Example 3.2.3](#). Using the coordinates from [Exercise 3.3.10](#), we have left-invariant vector fields on $\mathrm{Aff}(\mathbb{R})$ given by $u \frac{\partial}{\partial u}$ and $u \frac{\partial}{\partial v}$. Therefore, for any smooth function $h: \mathrm{Aff}(\mathbb{R}) \rightarrow \mathbb{R}$,

$$\left[u \frac{\partial}{\partial u}, u \frac{\partial}{\partial v} \right](h) = u \frac{\partial}{\partial u} \left(u \frac{\partial}{\partial v} h \right) - u \frac{\partial}{\partial v} \left(u \frac{\partial}{\partial u} h \right) = u \frac{\partial h}{\partial v} + u^2 \frac{\partial^2 h}{\partial u \partial v} - u^2 \frac{\partial^2 h}{\partial v \partial u} = u \frac{\partial}{\partial v}(h).$$

That is, $[u \frac{\partial}{\partial u}, u \frac{\partial}{\partial v}] = u \frac{\partial}{\partial v}$. On the other hand, $df(u \frac{\partial}{\partial u})$ and $df(u \frac{\partial}{\partial v})$ correspond to the matrices $A = \begin{bmatrix} 1 & 0 \\ 0 & 0 \end{bmatrix}$ and $B = \begin{bmatrix} 0 & 1 \\ 0 & 0 \end{bmatrix}$ from [Example 3.3.9](#), and we saw there that $[A, B] = B$. Writing things out in terms of f we have

$$\left[df\left(u \frac{\partial}{\partial u}\right), df\left(u \frac{\partial}{\partial v}\right) \right] = [A, B] = B = df\left(u \frac{\partial}{\partial v}\right) = df\left(\left[u \frac{\partial}{\partial u}, u \frac{\partial}{\partial v}\right]\right),$$

so $df: \mathfrak{aff}(\mathbb{R}) \rightarrow \mathrm{gl}_2(\mathbb{R})$ is a Lie algebra homomorphism.

[Proposition 3.6.3](#) shows that Lie group homomorphisms induce Lie algebra homomorphisms, and so the expression $d\phi = \psi$ in [Theorem 3.6.1](#) makes sense. In order to prove the rest of [Theorem 3.6.1](#), we'll need to make an extended detour into the theory of distributions (which are also useful for other things).

3.7 Distributions and the Frobenius Theorem

Think back to [Section 1.2](#) when we defined tangent spaces and the tangent bundle. The upshot of that construction ([Definition 1.2.1](#)) was that we assigned a vector space to each point on the manifold, and that the elements of each tangent space were all possible velocities of smooth curves passing through the point.

Now, in a variety of situations, we may be particularly interested in only certain types of curves passing through each point.

Example 3.7.1. Consider a homogeneous, first-order differential equation of the form $F(x, z, z') = 0$, where $z = z(x)$ is a function of x . Solutions correspond to curves $x \mapsto (x, z(x), z'(x))$, which we can think of as living in $\mathbb{R}^3 = \mathbb{R}^2 \times \mathbb{R}$ with coordinates (x, z, p) . But such curves cannot have arbitrary velocities (or tangent vectors): specifically, $dz = z' dx$, so the tangents to the curve must always lie in the kernel of the 1-form

$$\alpha = dz - p dx.$$

This 1-form is an example of a *contact form*, first introduced in the late 19th century by Sophus Lie [21]. He called this a *contact element* or a *line element* because $dz - p dx = 0$ can be interpreted as the differential form of the equation of a line of slope p passing through the point $(x, z) \in \mathbb{R}^2$.

Exercise 3.7.2. Show that $\ker \alpha = \text{span} \left\{ \frac{\partial}{\partial x} + p \frac{\partial}{\partial z}, \frac{\partial}{\partial p} \right\}$.

Therefore, $\ker \alpha$ determines a 2-dimensional subspace of each tangent space, visualized in Figure 3.1.



Figure 3.1: A visualization of the plane field $\ker \alpha = \text{span} \left\{ \frac{\partial}{\partial x} + p \frac{\partial}{\partial z}, \frac{\partial}{\partial p} \right\}$.

Solutions to differential equations of the form $F(x, z, z') = 0$ produce curves in (x, z, p) -space which are everywhere tangent to the plane determined by $\ker \alpha$; these are often called Legendrian curves. This formalism is the basis of an area of mathematics called *contact geometry* or *contact topology*, which has recently been very useful in the study of 3-manifolds, but which historically was developed because of its relevance to thermodynamics and geometric optics; see Geiges's historical survey [12] for an introduction to this area.

The plane field described above and illustrated in Figure 3.1 is an example of a *distribution*:

Definition 3.7.3. A d -dimensional *distribution* on a manifold M is a smooth choice of d -dimensional subspace $\mathcal{D}(p) \subset T_p M$ for each $p \in M$. Equivalently, \mathcal{D} is a sub-bundle of the tangent bundle TM . This means that at each point $p \in M$, there exists a neighborhood U of p and vector fields X_1, \dots, X_d so that $\mathcal{D}(p) = \text{span}\{X_1(q), \dots, X_d(q)\}$ for any $q \in U$.

A vector field $X \in \mathfrak{X}(M)$ lies in \mathcal{D} if $X(p) \in \mathcal{D}(p)$ for all $p \in M$; by slight abuse of notation, we denote this as $X \in \mathcal{D}$. The distribution is *involutive* if $[X, Y] \in \mathcal{D}$ for all $X, Y \in \mathcal{D}$.

Example 3.7.4. Let $\mathcal{D} = \text{span} \left\{ \frac{\partial}{\partial x_{i_1}}, \dots, \frac{\partial}{\partial x_{i_k}} \right\}$ for some $1 \leq i_1 < \dots < i_k \leq n$. Then \mathcal{D} is a k -dimensional distribution on \mathbb{R}^n . Moreover, for any $1 \leq r, s \leq k$,

$$\left[\frac{\partial}{\partial x_{i_r}}, \frac{\partial}{\partial x_{i_s}} \right] = \frac{\partial^2}{\partial x_{i_r} \partial x_{i_s}} - \frac{\partial^2}{\partial x_{i_s} \partial x_{i_r}} = 0$$

since mixed partials commute, so \mathcal{D} is involutive.

Example 3.7.5. Let $\xi = \text{span} \left\{ \frac{\partial}{\partial x} + p \frac{\partial}{\partial z}, \frac{\partial}{\partial p} \right\} = \ker(dz - p dx)$ be the 2-dimensional distribution on \mathbb{R}^3 from Example 3.7.1. Then for any $f \in C^\infty(\mathbb{R}^3)$,

$$\begin{aligned} \left[\frac{\partial}{\partial x} + p \frac{\partial}{\partial z}, \frac{\partial}{\partial p} \right] (f) &= \left(\frac{\partial}{\partial x} + p \frac{\partial}{\partial z} \right) \left(\frac{\partial f}{\partial p} \right) - \frac{\partial}{\partial p} \left(\frac{\partial f}{\partial x} + p \frac{\partial f}{\partial z} \right) \\ &= \frac{\partial^2 f}{\partial x \partial p} + p \frac{\partial^2 f}{\partial z \partial p} - \frac{\partial^2 f}{\partial p \partial x} - \frac{\partial f}{\partial z} - p \frac{\partial^2 f}{\partial p \partial z} \\ &= -\frac{\partial f}{\partial z}, \end{aligned}$$

so $\left[\frac{\partial}{\partial x} + p \frac{\partial}{\partial z}, \frac{\partial}{\partial p} \right] = -\frac{\partial}{\partial z} \notin \xi$, so ξ is not involutive. ξ is called the *standard contact structure* on \mathbb{R}^3 .

Example 3.7.6. Recall the left-invariant vector fields X, Y, Z on S^3 from Section 1.8. Let $\zeta = \text{span}\{Y, Z\} = \ker \alpha$, where α is the 1-form dual to X . ζ is a 2-dimensional distribution on S^3 . We saw in Section 1.8 that $[Y, Z] = 2X \notin \zeta$, so ζ is also not involutive. ζ is called the *standard tight contact structure* on S^3 .

Definition 3.7.7. The image of a smooth embedding $i: N \rightarrow M$ is an *integral manifold* of a distribution \mathcal{D} on M if $(di)_p(T_p N) = \mathcal{D}(i(p))$ for all $p \in N$. In other words, N has the same dimension as \mathcal{D} and $i(N)$ is everywhere tangent to \mathcal{D} .

Example 3.7.8. Consider $\mathcal{D} = \text{span} \left\{ -y \frac{\partial}{\partial x} + x \frac{\partial}{\partial y}, -z \frac{\partial}{\partial x} + x \frac{\partial}{\partial z} \right\}$ on $\mathbb{R}^3 - \{\vec{0}\}$. Notice, first of all, that \mathcal{D} is involutive since

$$\begin{aligned} \left[-y \frac{\partial}{\partial x} + x \frac{\partial}{\partial y}, -z \frac{\partial}{\partial x} + x \frac{\partial}{\partial z} \right] &= \left(-y \frac{\partial}{\partial x} + x \frac{\partial}{\partial y} \right) \left(-z \frac{\partial}{\partial x} + x \frac{\partial}{\partial z} \right) - \left(-z \frac{\partial}{\partial x} + x \frac{\partial}{\partial z} \right) \left(-y \frac{\partial}{\partial x} + x \frac{\partial}{\partial y} \right) \\ &= yz \frac{\partial^2}{\partial x^2} - xz \frac{\partial^2}{\partial y \partial x} - y \frac{\partial}{\partial z} - xy \frac{\partial^2}{\partial x \partial z} + x^2 \frac{\partial^2}{\partial y \partial z} - yz \frac{\partial^2}{\partial x^2} + z \frac{\partial}{\partial y} + xz \frac{\partial^2}{\partial x \partial y} + xy \frac{\partial^2}{\partial z \partial x} - x^2 \frac{\partial^2}{\partial z \partial y} \\ &= -y \frac{\partial}{\partial z} + z \frac{\partial}{\partial y} = -\frac{y}{x} \left(-z \frac{\partial}{\partial x} + x \frac{\partial}{\partial z} \right) + \frac{z}{x} \left(-y \frac{\partial}{\partial x} + x \frac{\partial}{\partial y} \right) \in \mathcal{D}. \end{aligned}$$

Also, since the cross product

$$\left(-y \frac{\partial}{\partial x} + x \frac{\partial}{\partial y} \right) \times \left(-z \frac{\partial}{\partial x} + x \frac{\partial}{\partial z} \right) = x \left(x \frac{\partial}{\partial x} + y \frac{\partial}{\partial y} + z \frac{\partial}{\partial z} \right)$$

is parallel to the radial field $x \frac{\partial}{\partial x} + y \frac{\partial}{\partial y} + z \frac{\partial}{\partial z}$, we see that \mathcal{D} is everywhere tangent to the concentric spheres centered at the origin, which are integral manifolds for \mathcal{D} .

The algebraic condition of involutivity is closely related to the geometric condition of having an integral manifold:

Proposition 3.7.9. If \mathcal{D} is a smooth distribution on M so that at each $p \in M$ there is an integral manifold of \mathcal{D} passing through p , then \mathcal{D} is involutive.

Proof. If $X, Y \in \mathcal{D}$, then at each $q = i(p) \in i(N)$, we have that $X(q) = (di)_p \tilde{X}(p)$ and $Y(q) = (di)_p \tilde{Y}(p)$ for some $\tilde{X}, \tilde{Y} \in \mathfrak{X}(N)$. Since $[\tilde{X}, \tilde{Y}](p) \in T_p N$, we see that

$$[X, Y](q) = [(di)_p \tilde{X}(p), (di)_p \tilde{Y}(p)](i(p)) = (di)_p [\tilde{X}, \tilde{Y}] \in (di)_p(T_p N) = \mathcal{D}(q)$$

using Lemma 3.3.8. So \mathcal{D} is involutive. \square

The much more amazing fact is that the converse is true, so we can determine existence integral manifolds by checking involutivity. This is the content of:

Theorem 3.7.10 (Frobenius Theorem). *Let \mathcal{D} be a k -dimensional involutive distribution on a manifold M . Then for each $p \in M$ there is an integral manifold of \mathcal{D} passing through p .*

The idea behind the theorem is the following: suppose \mathcal{D} is (locally) spanned by the vector fields V_1, \dots, V_k . Remember that the V_i are really first-order differential operators on M , so the set $\{V_1, \dots, V_k\}$ determines a first-order homogeneous system of PDEs

$$V_1 u = 0, \dots, V_k u = 0.$$

If $u: M \rightarrow \mathbb{R}$ is a solution to this system of PDEs, then the level sets of u are tangent to $\text{span}\{V_1, \dots, V_k\}$. Frobenius' original question was: when are there solutions u_1, \dots, u_{n-k} so that the gradients $\nabla u_1, \dots, \nabla u_{n-k}$ are linearly independent? (There certainly can't be more than $n - k$ solutions, since the gradients will be normal to the level sets and the orthogonal complement of $\text{span}\{V_1, \dots, V_k\}$ is $(n - k)$ -dimensional.)

When there exist such u_1, \dots, u_{n-k} , then the level sets of the product map $(u_1, \dots, u_{n-k}): M \rightarrow \mathbb{R}^{n-k}$ will necessarily be integral manifolds of \mathcal{D} .

Proof of Theorem 3.7.10. I want to show that, for all $p \in M$, there exists a coordinate chart (U, ϕ) so that:

- (i) $\vec{0} \in U \subset \mathbb{R}^n$
- (ii) $\phi(\vec{0}) = p$
- (iii) For $\vec{a} = (a_1, \dots, a_n) \in U$, the set $\{q \in M : \phi^{-1}(q)_{k+1} = a_{k+1}, \dots, \phi^{-1}(q)_n = a_n\}$ is an integral manifold of \mathcal{D} .

Since this is a local argument, we may as well assume $M \subset \mathbb{R}^n$ is an open set, $p = \vec{0}$, and, after rotation, $\mathcal{D}(p) = \mathcal{D}(\vec{0}) = \text{span} \left\{ \frac{\partial}{\partial x_1}, \dots, \frac{\partial}{\partial x_k} \right\}$ is the k -plane spanned by the first k coordinate directions.

Let $\pi: \mathbb{R}^n \rightarrow \mathbb{R}^k$ be projection onto the first k coordinates, so that $(d\pi)_{\vec{0}}|_{\mathcal{D}}: \mathcal{D}(\vec{0}) \rightarrow T_{\vec{0}} \mathbb{R}^k \cong \mathbb{R}^k$ is an isomorphism, and hence $d\pi|_{\mathcal{D}}$ is injective on some neighborhood U of $\vec{0}$.

Therefore, in U there exist unique $V_1, \dots, V_k \in \mathcal{D}$ so that $(d\pi)_q V_i = \left. \frac{\partial}{\partial x_i} \right|_{\pi(q)}$ for all $q \in U$ and for all $i \in \{1, \dots, k\}$.

Since \mathcal{D} is involutive, $[V_i, V_j] \in \mathcal{D}$ for all $1 \leq i, j \leq k$. On the other hand,

$$d\pi[V_i, V_j] = [d\pi V_i, d\pi V_j] = \left[\left. \frac{\partial}{\partial x_i} \right|_{\pi(q)}, \left. \frac{\partial}{\partial x_j} \right|_{\pi(q)} \right] = 0,$$

so $[V_i, V_j] = 0$ since $d\pi$ is an isomorphism.

But then we can find local coordinates y_1, \dots, y_n so that $V_i = \frac{\partial}{\partial y_i}$ on U and the sets

$$\{y \in U : y_{k+1} = a_{k+1}, \dots, y_n = a_n\}$$

are integral manifolds of \mathcal{D} . □

Example 3.7.11. Let $M = \mathbb{R}^2 - \{\vec{0}\}$, let $X(u, v) = -v \frac{\partial}{\partial x} + u \frac{\partial}{\partial y}$, and let $\mathcal{D} = \text{span}\{X\}$. \mathcal{D} is certainly involutive since $[X, X] = 0$ by anticommutativity of the Lie bracket, so there must be (local) integral manifolds through each point in M . Of course, this example is easy: the integral manifolds are just concentric circles, as we see in [Figure 3.2](#).

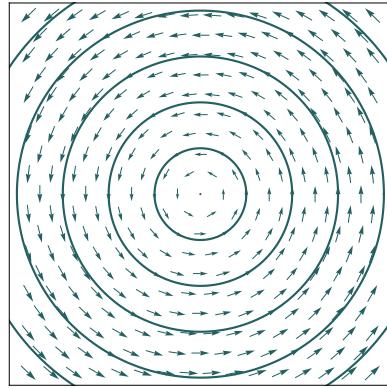


Figure 3.2: The vector field $X(u, v) = -v \frac{\partial}{\partial x} + u \frac{\partial}{\partial y}$ together with some integral curves, which are concentric circles.

In fact, anticommutativity of the Lie bracket implies that *all* one-dimensional distributions are involutive, and hence integrable. Said another way: given any line field on a manifold and any point in the manifold, there is a curve passing through the point whose velocity vectors are everywhere tangent to the line field.

Example 3.7.12. Consider the standard contact structure $\xi = \text{span} \left\{ \frac{\partial}{\partial x} + p \frac{\partial}{\partial z}, \frac{\partial}{\partial p} \right\} = \ker(dz - p dx)$ on \mathbb{R}^3 defined in [Example 3.7.1](#). We saw in [Example 3.7.5](#) that ξ is not involutive, so there are no surfaces in \mathbb{R}^3 which are tangent to ξ in a neighborhood of any point. In other words, ξ is a *totally non-integrable plane field*, which is typically how contact structures on 3-manifolds are defined.

The Frobenius theorem we've proved is a local result (after all, there's no guarantee that the integral manifold we get has diameter $> \epsilon$), but it turns out that the local pieces glue together nicely:

Theorem 3.7.13 (Global Frobenius Theorem). *Let \mathcal{D} be an involutive distribution on M . Then M is foliated by a (disconnected) integral manifold for \mathcal{D} . The components of this foliation are called the maximal integral manifolds for \mathcal{D} .*

Example 3.7.14. The foliation of $\mathbb{R}^2 - \{\vec{0}\}$ corresponding to the distribution $\mathcal{D} = \text{span} \left\{ -y \frac{\partial}{\partial x} + x \frac{\partial}{\partial y} \right\}$ from [Example 3.7.11](#) is precisely the disjoint union of concentric circles shown in [Figure 3.3](#).

Example 3.7.15. The foliation of \mathbb{R}^3 corresponding to the involutive distribution $\mathcal{D} = \text{span} \left\{ \frac{\partial}{\partial x}, \frac{\partial}{\partial y} \right\}$ is the collection of planes parallel to the xy -plane, as shown in [Figure 3.4](#).



Figure 3.3: A foliation of the punctured plane by concentric circles.

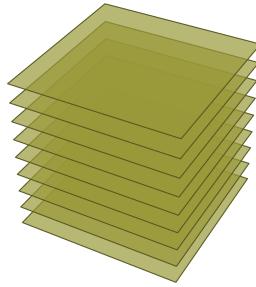


Figure 3.4: A foliation of \mathbb{R}^3 by parallel planes.

3.8 Distributions and Differential Ideals

3.8.1 Linear Subspaces and Annihilators

So far, we've connected a geometric property of distributions (having integral manifolds) to an algebraic property in terms of vector fields tangent to the distribution (involutivity). Roughly speaking, we should expect algebraic properties of tangent vectors to carry over to algebraic properties of differential forms, so that's what we pursue next.

First, let's just focus on $\mathcal{D}(p) \subset T_p M$ for a particular point $p \in M$. This is just a subspace of a vector space. Of course, one way to describe a subspace is to describe the elements, for example by describing it as the span of a collection of vectors. Thinking dually, we could also describe the subspace in terms of the linear functionals which vanish on it.³

Example 3.8.1. Think about the challenge of describing a 2-dimensional subspace $S \subset \mathbb{R}^3$. One way to do so is to give a basis for S ; i.e., $S = \text{span}\{\vec{u}, \vec{v}\}$ for some particular $\vec{u}, \vec{v} \in \mathbb{R}^3$. This is very convenient if we want to do computations involving vectors which we already know are in S ; for example, if $\vec{w} \in S$, then $\vec{w} = a\vec{u} + b\vec{v}$, and then for any linear transformation T we know that $T(\vec{w}) = aT(\vec{u}) + bT(\vec{v})$, so we only really need to know $T(\vec{u})$ and $T(\vec{v})$ to describe what T does to anything in S .

On the other hand, this description is not so convenient if we don't already know \vec{w} is in S . At the most

³Thought of another way, we can always think of a subspace as the kernel of some linear map. This is analogous to how in algebra we can think of normal subgroups as kernels of group homomorphisms.

basic level, we might be trying to determine whether some vector $\vec{w} \in \mathbb{R}^3$ is in S . We then have to determine whether the linear equation $a\vec{u} + b\vec{v} = \vec{w}$ has a solution. Admittedly, this is not a particularly challenging problem, but it might still be moderately annoying.

For this kind of problem, it is much nicer to describe S in terms of linear functionals. For example, if we happen to know a vector \vec{n} which is perpendicular to S (for example, $\vec{n} = \vec{u} \times \vec{v}$, if we happen to have a basis $\{\vec{u}, \vec{v}\}$), then we could define the linear functional $\alpha: \mathbb{R}^3 \rightarrow \mathbb{R}$ by $\alpha(\vec{w}) = \vec{n} \cdot \vec{w}$, and we see that $S = \ker \alpha$. Then it's easy to determine whether any given \vec{w} is in S : just compute $\alpha(\vec{w})$ and see if it's zero. Of course, describing planes in \mathbb{R}^3 in terms of their (unit) normals is a very common and useful approach.

In [Example 3.8.1](#), I used the fact that I had a preferred inner product on \mathbb{R}^3 , but this isn't really necessary.

Definition 3.8.2. Let S be a subspace of a vector space V . The *annihilator* of S , denoted S^0 , is defined by

$$S^0 := \{\phi \in V^* : \phi(v) = 0 \text{ for all } v \in S\}.$$

The annihilator is a subset of the dual space V^* , and in fact it is easy to show it is a subspace:

Proposition 3.8.3. *If S is a subspace of a vector space V , then the annihilator S^0 is a subspace of V^* .*

Moreover, as you might guess from the \mathbb{R}^3 example, S and S^0 have complementary dimensions (here I'm assuming $\dim V$ is finite):

$$\dim S + \dim S^0 = \dim V.$$

In fact, if V has an inner product which we use to define an explicit isomorphism with V^* , then $S^0 \subset V^*$ will map to the orthogonal complement of S under this isomorphism.

Annihilators also interact well with dual maps:

Proposition 3.8.4. *Suppose V and W are finite-dimensional vector spaces and $T: V \rightarrow W$ is linear. Let $T^*: W^* \rightarrow V^*$ be the corresponding dual map. Then*

- (i) $\ker T^* = (\text{im } T)^0$
- (ii) $\dim(\ker T^*) = \dim(\ker T) + \dim W - \dim V$
- (iii) $\text{im } T^* = (\ker T)^0$
- (iv) $\dim(\text{im } T^*) = \dim(\text{im } T)$

For a proof, see [3, §3.F].

Notice, in particular, that the choice of a basis v_1, \dots, v_k for a subspace $S \subset V$ induces a linear map $B: \mathbb{R}^k \rightarrow V$ defined by

$$B(e_i) = v_i,$$

where e_1, \dots, e_k is the standard basis for \mathbb{R}^k (or really any basis). Of course, $S = \text{im}(B)$, so [Proposition 3.8.4\(i\)](#) implies that

$$S^0 = (\text{im } B)^0 = \ker B^*,$$

where $B^*: V^* \rightarrow (\mathbb{R}^k)^*$ is the dual map $B^*(\sigma)(v) = \sigma(B(v))$. So this gives a (semi-)explicit description of the linear functionals which vanish on S .

In the context of distributions on manifolds, $\mathcal{D}(p)$ is a subspace of $T_p M$, and the space of linear functionals $(T_p M)^* = \bigwedge^1 (T_p M)^*$ is just the 1-forms on M at p . So, more generally, we can talk about the differential forms on M which vanish on \mathcal{D} . In fact, we've already seen this in [Example 3.7.1](#), where we described the standard contact structure on \mathbb{R}^3 as the kernel of a 1-form, namely $\ker(dz - p dx)$.

3.8.2 Differential Ideals

Now we formalize our approach to describing distributions as kernels of differential forms.

Definition 3.8.5. Suppose \mathcal{D} is a d -dimensional distribution on an n -dimensional manifold M and define $\mathcal{I}(\mathcal{D}) \subset \Omega^*(M)$ to be the ring of differential forms which annihilate \mathcal{D} :

$$\mathcal{I}(\mathcal{D}) := \bigcup_k \left\{ \omega \in \Omega^k(M) : \omega(X_1, \dots, X_k) = 0 \text{ for all } X_1, \dots, X_k \in \mathcal{D} \right\}.$$

If $\omega_1, \omega_2 \in \mathcal{I}(\mathcal{D})$, then $\omega_1 + \omega_2 \in \mathcal{I}(\mathcal{D})$ and $\omega_1 \wedge \omega_2 \in \mathcal{I}(\mathcal{D})$, so this really is a ring.

In fact, if $\omega \in \mathcal{I}(\mathcal{D}) \cap \Omega^k(M)$ and $\eta \in \Omega^\ell(M)$, then

$$\omega \wedge \eta(X_1, \dots, X_{k+\ell}) = \sum_{\sigma \in S(k, \ell)} \text{sgn}(\sigma) \omega(X_{\sigma(1)}, \dots, X_{\sigma(k)}) \eta(X_{\sigma(k+1)}, \dots, X_{\sigma(k+\ell)}) = 0,$$

so $\omega \wedge \eta \in \mathcal{I}(\mathcal{D})$, which means that $\mathcal{I}(\mathcal{D})$ is actually an ideal, usually called either the *annihilator ideal* of \mathcal{D} or just the *annihilator* of \mathcal{D} .

We've seen before in the proof of [Theorem 3.7.10](#) that any point $p \in M$ is contained in a coordinate chart (U, ϕ) with local coordinates x_1, \dots, x_n so that $\mathcal{D}(p) = \text{span} \left\{ \frac{\partial}{\partial x_1}, \dots, \frac{\partial}{\partial x_d} \right\}$. If ξ_1, \dots, ξ_n are the dual 1-forms (i.e., $\xi_i = dx_i$), then ξ_{d+1}, \dots, ξ_n are independent and generate $\mathcal{I}(\mathcal{D})$, which proves:

Lemma 3.8.6. *For any d -dimensional distribution \mathcal{D} , the annihilator ideal $\mathcal{I}(\mathcal{D})$ is locally generated by $n - d$ independent 1-forms ξ_{d+1}, \dots, ξ_n .*

On the other hand:

Lemma 3.8.7. *If $\mathcal{I} \subset \Omega^*(M)$ is an ideal locally generated by $n - d$ independent 1-forms, then there is a d -dimensional distribution \mathcal{D} on M so that $\mathcal{I} = \mathcal{I}(\mathcal{D})$.*

Proof. Let $\omega_{d+1}, \dots, \omega_n$ be the independent 1-forms generating \mathcal{I} in a neighborhood of $p \in M$. Then define $\mathcal{D}(p) \subset T_p M$ to be the d -dimensional subspace annihilated by the linear functionals $(\omega_{d+1})_p, \dots, (\omega_n)_p$. Then $\mathcal{D} = \bigcup_{p \in M} \mathcal{D}(p)$ is the desired distribution. \square

Now we can give an alternative version of the Frobenius theorem in terms of annihilator ideals:

Theorem 3.8.8 (Frobenius Theorem, version 2). *A distribution \mathcal{D} on M is involutive (and hence integrable) if and only if $\mathcal{I}(\mathcal{D})$ is closed under taking exterior derivatives: $d(\mathcal{I}(\mathcal{D})) \subseteq \mathcal{I}(\mathcal{D})$.*

Definition 3.8.9. An ideal $\mathcal{I} \subset \Omega^*(M)$ is called a *differential ideal* if $d(\mathcal{I}) \subseteq \mathcal{I}$.

So a restatement of [Theorem 3.8.8](#) is that a distribution is involutive if and only if its annihilator ideal is a differential ideal.

Proof of Theorem 3.8.8. Locally choose ξ_1, \dots, ξ_n that $\text{span} (T_p M)^* = \bigwedge^1 ((T_p M)^*)$ so that ξ_{d+1}, \dots, ξ_n generate $\mathcal{I}(\mathcal{D})$ in a neighborhood of $p \in M$. Let X_1, \dots, X_n be the dual vector fields so that $\xi_i(X_j) = \delta_{ij}$. Then X_1, \dots, X_d span \mathcal{D} and \mathcal{D} is involutive if and only if $[X_i, X_j] \in \mathcal{D}$ for all $1 \leq i, j \leq d$.

But now we get to use Cartan's Magic Formula ([Theorem 2.7.5](#)), which says that $\mathcal{L}_X\omega = \iota_X d\omega + d\iota_X\omega$. Then for any 1-form ω and any $X, Y \in \mathfrak{X}(M)$, this implies that

$$\begin{aligned} d\omega(X, Y) &= (\iota_X d\omega)(Y) = (\mathcal{L}_X\omega)(Y) - (d\iota_X\omega)(Y) \\ &= \mathcal{L}_X(\omega(Y)) - \omega(\mathcal{L}_X(Y)) - (d\omega(X))(Y) \\ &= X(\omega(Y)) - \omega([X, Y]) - Y(\omega(X)), \end{aligned}$$

since $\omega(X)$ and $\omega(Y)$ are functions.

We record this as a lemma:

Lemma 3.8.10. *Suppose $\omega \in \Omega^1(M)$ and $X, Y \in \mathfrak{X}(M)$. Then*

$$d\omega(X, Y) = X(\omega(Y)) - Y(\omega(X)) - \omega([X, Y]).$$

This implies that, if $1 \leq i, j \leq d$ and $d+1 \leq k \leq n$,

$$d\xi_k(X_i, X_j) = X_i(\xi_k(X_j)) - X_j(\xi_k(X_i)) - \xi_k([X_i, X_j]) = X_i(0) - X_j(0) - \xi_k([X_i, X_j]) = -\xi_k([X_i, X_j]).$$

This is zero for all such i, j, k if and only if \mathcal{D} is involutive. So $\mathcal{I}(\mathcal{D})$ is a differential ideal if and only if \mathcal{D} is involutive. \square

Example 3.8.11. Recall yet again [Section 1.8](#) and our left-invariant vector fields X, Y, Z and dual left-invariant 1-forms α, β, γ on S^3 . Let $\zeta = \text{span}\{Y, Z\} = \ker \alpha$. Recall from [Example 3.7.6](#) that $[Y, Z] = 2X \notin \zeta$, so ζ is not involutive, which by the original Frobenius Theorem ([Theorem 3.7.10](#)) implies that ζ is not integrable. We can also see this using [Theorem 3.8.8](#): certainly $\alpha \in \mathcal{I}(\zeta)$, but as we saw in [Example 2.7.6](#),

$$d\alpha = -2\beta \wedge \gamma$$

which is not in $\mathcal{I}(\zeta)$ since $-2\beta \wedge \gamma(Y, Z) = -2 \neq 0$, so $\mathcal{I}(\zeta)$ is not a differential ideal.

We (finally) have all the tools in place to start using this distribution machinery to prove [Theorem 3.6.1](#), which said that Lie algebra homomorphisms induce Lie group homomorphisms.

Here's an intermediate result:

Theorem 3.8.12. *Let G be a connected Lie group and let $f_1, f_2: G \rightarrow H$ be Lie group homomorphisms so that the Lie algebra homomorphisms $df_1, df_2: \mathfrak{g} \rightarrow \mathfrak{h}$ are identical. Then $f_1 \equiv f_2$.*

Let's start trying to prove this and see where we run into trouble. Since $df_1 = df_2$ as maps $\mathfrak{g} \rightarrow \mathfrak{h}$, the dual maps $f_1^* = f_2^*$ as maps $\mathfrak{h}^* \rightarrow \mathfrak{g}^*$, where we identify \mathfrak{g}^* and \mathfrak{h}^* with the left-invariant 1-forms on G and H , respectively.⁴

Suppose $\omega_1, \dots, \omega_n$ is a basis for the left-invariant forms on H . Then we know that $f_1^*\omega_i = f_2^*\omega_i$ for all $i = 1, \dots, n$, and the question is whether this is enough to show that $f_1 \equiv f_2$.

Think of this as a special case of a more general problem:

⁴Notice that a Lie group homomorphism f really does pull back left-invariant forms to left-invariant forms: for ω left-invariant on H and $g \in G$,

$$L_g^* f^* \omega = (f \circ L_g)^* \omega = (L_{f(g)} \circ f)^* \omega = f^* L_{f(g)}^* \omega = f^* \omega,$$

where the second equality follows by the same argument as in the proof of [Proposition 3.6.3](#).

Problem 3.8.13. Suppose we have manifolds M and N of dimensions m and n , respectively, and we have a basis $\omega_1, \dots, \omega_n$ for the 1-forms on N , as well as a preferred collection $\alpha_1, \dots, \alpha_m$ of 1-forms on M . Is there a (preferably unique) map $f: M \rightarrow N$ so that

$$f^*\omega_i = \alpha_i$$

for all $i = 1, \dots, n$?

Example 3.8.14. Let $M = \mathbb{R}$, let $N = \mathbb{R}_{>0}$ be the positive reals, let $\omega = \frac{1}{y} dy \in \Omega^1(N)$, and let $\alpha = dx \in \Omega^1(M)$. If there is some $f: M \rightarrow N$ so that $f^*\omega = \alpha$, then it must be the case that, for each $r \in \mathbb{R}$,

$$1 = dx_r \left(\frac{\partial}{\partial x} \right) = \alpha_r \left(\frac{\partial}{\partial x} \right) = (f^*\omega)_r \left(\frac{\partial}{\partial x} \right) = \omega_{f(r)} \left(df_r \frac{\partial}{\partial x} \right) = \omega_{f(r)} ((f \circ \gamma)'(0)),$$

where $\gamma(0) = r$ and $\gamma'(0) = \frac{\partial}{\partial x}$; for example, $\gamma(t) = r + t$. In this case, $(f \circ \gamma)(t) = f(r + t)$, so $(f \circ \gamma)'(0) = f'(r) \frac{\partial}{\partial y}$, and hence we see that

$$1 = \omega_{f(r)} ((f \circ \gamma)'(0)) = \omega_{f(r)} \left(f'(r) \frac{\partial}{\partial y} \right) = \frac{1}{f(r)} dy \left(f'(r) \frac{\partial}{\partial y} \right) = \frac{1}{f(r)} f'(r).$$

This can be solved, for example with $f_c(r) = ce^r$ for any positive constant $c > 0$. If we specify that $f(0) = 1$, then that fixes $c = 1$ and we have the unique solution $f(r) = e^r$.

To make this more geometric, for each $c > 0$ consider the graph of f_c , shown in Figure 3.5:

$$\Gamma_c := \{(r, f_c(r)) : r \in \mathbb{R}\} = \{(r, ce^r) : r \in \mathbb{R}\}.$$

Notice that the tangent line to a point $(r, ce^r) \in \Gamma_c$ is spanned by the vector $(1, ce^r) = \frac{\partial}{\partial x} + ce^r \frac{\partial}{\partial y}$. If we

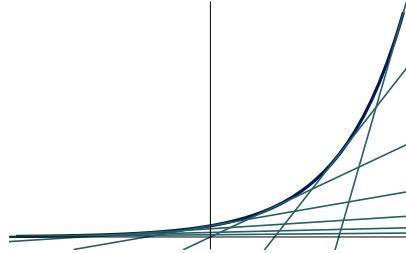


Figure 3.5: The graph of the function $f_c(r) = ce^r$, shown with its family of tangent lines.

write (r, ce^r) as (x, y) , then this vector is just $\frac{\partial}{\partial x} + y \frac{\partial}{\partial y}$, which defines a non-vanishing vector field (and hence a line field or 1-dimensional distribution) on all of the upper half-plane $\mathbb{R} \times \mathbb{R}_{>0}$. And for each c the graph Γ_c is an integral manifold of this distribution. See Figure 3.6.

Finally, we see that the annihilator ideal of this distribution is generated by the 1-form dual to $\frac{\partial}{\partial x} + y \frac{\partial}{\partial y}$, namely $\mu = dx - \frac{1}{y} dy$. But now this is a form we could have constructed on $M \times N = \mathbb{R} \times \mathbb{R}_{>0}$ without knowing anything at all about f : if π_1 and π_2 are the obvious projections onto the first and second factors, then $\mu = \pi_1^*\alpha - \pi_2^*\omega$.

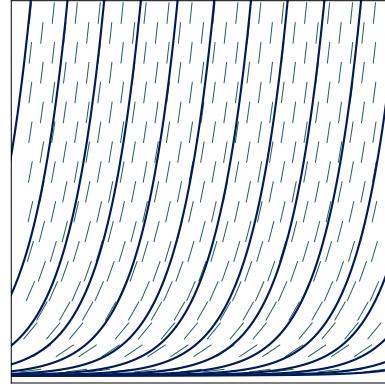


Figure 3.6: The integral manifolds of the line field $\frac{\partial}{\partial x} + y \frac{\partial}{\partial y}$ on the upper half-plane.

For arbitrary M and N with α_i and ω_i given as above, define $\mu_i \in \Omega^1(M \times N)$ by

$$\mu_i := \pi_1^* \alpha_i - \pi_2^* \omega_i$$

and let \mathcal{I} be the ideal generated by the μ_i .

This gives a strategy for solving [Problem 3.8.13](#): suppose there were a map satisfying our condition $f^* \omega_i = \alpha_i$ for all i , and let $\Gamma: M \rightarrow M \times N$ be the graph of f : $\Gamma(x) = (x, f(x))$ for all $x \in M$. Then I claim that Γ is an integral manifold of \mathcal{I} (really, of the distribution \mathcal{D} so that $\mathcal{I} = \mathcal{I}(\mathcal{D})$).

To see this, it suffices to show that $\Gamma^* \mu_i = 0$ for all i . Since $\pi_1 \circ \Gamma = \text{id}_M$ and $\pi_2 \circ \Gamma = f$,

$$\Gamma^* \mu_i = \Gamma^*(\pi_1^* \alpha_i - \pi_2^* \omega_i) = \Gamma^*(\pi_1^* f^* \omega_i - \pi_2^* \omega_i) = (\pi_1 \circ \Gamma)^* f^* \omega_i - (\pi_2 \circ \Gamma)^* \omega_i = f^* \omega_i - f^* \omega_i = 0.$$

So this says that if f exists, its graph is an integral manifold of the ideal \mathcal{I} , which of course means that \mathcal{I} must be a differential ideal (by [Theorem 3.8.8](#)). In other words, \mathcal{I} being a differential ideal is a necessary condition for such an f to exist.

3.9 Lie Group Homomorphisms are Determined by their Differentials

Let's recap and see where we are. We're working our way towards a proof of [Theorem 3.6.1](#), which (essentially) says that Lie group homomorphisms are uniquely determined by their differentials and that every Lie algebra homomorphism is the differential of some Lie group homomorphism.

To do that, we want to prove the intermediate result [Theorem 3.8.12](#), in which we're trying to show that if $f_1^*, f_2^*: \mathfrak{h}^* \rightarrow \mathfrak{g}^*$ have $f_1^* \omega_i = f_2^* \omega_i$ for some basis $\omega_1, \dots, \omega_n$ for left-invariant 1-forms on H , then $f_1 \equiv f_2$.

In turn, we saw this as a special case of [Problem 3.8.13](#), in which we have manifolds M and N , a basis $\omega_1, \dots, \omega_n$ for the 1-forms on N (which doesn't always exist, but does on any Lie group) and some preferred $\alpha_1, \dots, \alpha_n \in \Omega^1(M)$, and we would like to find a map $f: M \rightarrow N$ so that $f^* \omega_i = \alpha_i$ for all $i = 1, \dots, n$. We've seen that the ideal \mathcal{I} on $M \times N$ generated by the 1-forms $\mu_1 = \pi_1^* \alpha_1 - \pi_2^* \omega_1 \in \Omega^1(M \times N)$ must be a differential ideal if such an f exists. In fact, this is also a sufficient condition for the existence of such an f , which is essentially unique:

Theorem 3.9.1. If the ideal \mathcal{I} defined above is a differential ideal then, given $p \in M$ and $q \in N$, there exists a neighborhood U of p and a smooth map $f : M \rightarrow N$ so that $f(p) = q$ and $f^*\omega_i = \alpha_i$ for all $i = 1, \dots, n$. Moreover, f is unique relative to the choice of U .

Proof. Since \mathcal{I} is a differential ideal, Theorem 3.8.8 guarantees that there exists a maximal connected integral manifold S of \mathcal{I} through the point $(p, q) \in M \times N$. Now, the idea is to see that S is transverse to the M directions. If so, we can basically just define f to be $\pi_2 \circ \pi_1^{-1}$. Figure 3.7 shows the possibilities we're trying to rule out.

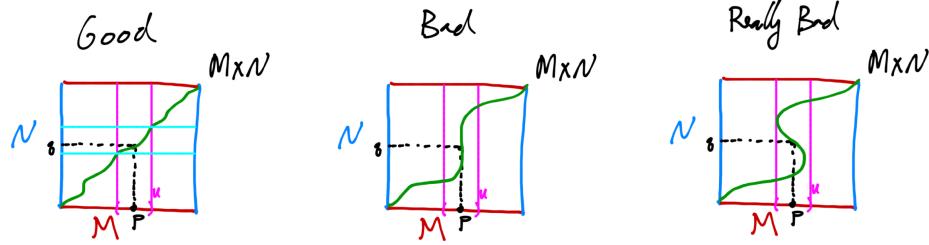


Figure 3.7: Three possibilities for the integral submanifold S of the ideal \mathcal{I} .

In particular, I claim that $d\pi_1|_S$ is injective. To see this, let $(r, s) \in S$ and let $v \in T_{(r,s)}S$. If $d\pi_1(v) = 0$, then

- (i) $\mu_i(v) = 0$ for all $i = 1, \dots, n$ since \mathcal{I} consists of all the forms which vanish on the tangent space to S .
- (ii) $\pi_1^*\alpha_i(v) = \alpha_i(d\pi_1 v) = \alpha_i(0) = 0$.

But then these two facts together imply that

$$\pi_2^*\omega_i(v) = \pi_1^*\alpha_i(v) - \mu_i(v) = 0 - 0 = 0$$

for all $i = 1, \dots, n$. Since the ω_i form a basis of 1-forms on N , this means that $d\pi_2(v) = 0$.

In turn, this implies that $v = 0$ since $T_{(r,s)}M \times N = T_r M \oplus T_s N$. Since $(r, s) \in S$ was arbitrary, we see that $d\pi_1|_S$ is injective, so the Inverse Function Theorem (Theorem 1.4.7) implies that $\pi_1|_S : S \rightarrow M$ is a local diffeomorphism since $\dim S = (m+n) - n = m = \dim M$. Therefore, there exists a neighborhood V of (p, q) in S and a neighborhood U of p in M so that $\pi_1|_V : V \rightarrow U$ is a diffeomorphism.

We're finally ready to define $f : U \rightarrow N$ by $f = \pi_2 \circ (\pi_1|_V)^{-1}$. Of course now we need to check that $f^*\omega_i = \alpha_i$ for all $i = 1, \dots, n$. To see this, pick $i \in \{1, \dots, n\}$ and $u \in T_p M$. Then

$$\begin{aligned} f^*\omega_i(u) &= \left(\pi_2 \circ (\pi_1|_V)^{-1} \right)^* \omega_i(u) = \left((\pi_1|_V)^{-1} \right)^* \pi_2^*\omega_i(u) \\ &= \pi_2^*\omega_i \left(d(\pi_1|_V)^{-1} u \right) = \pi_1^*\alpha_i \left(d(\pi_1|_V)^{-1} u \right) - \mu_i \left(d(\pi_1|_V)^{-1} u \right) \end{aligned}$$

by the definition of $\mu_i = \pi_1^*\alpha_i - \pi_2^*\omega_i$. The second term above is zero since $d(\pi_1|_V)^{-1} u \in T_{(r,s)}S$ and $\mathcal{I} = \mathcal{I}(S)$ is generated by the μ_i . Hence, we can continue the above computation to see

$$f^*\omega_i(u) = \pi_1^*\alpha_i \left(d(\pi_1|_V)^{-1} u \right) = \alpha_i \left(d\pi_1 d(\pi_1|_V)^{-1} u \right) = \alpha_i \left(d \left(\pi_1 \circ (\pi_1|_V)^{-1} \right) u \right) = \alpha_i(u)$$

since $\pi_1 \circ (\pi_1|_V)^{-1}$ is just the identity map. □

Exercise 3.9.2. Prove that the f defined in the above proof is unique.

Now let's apply this result to proving [Theorem 3.8.12](#), which I restate for convenience:

Theorem (Theorem 3.8.12). Let G be a connected Lie group and let $f_1, f_2: G \rightarrow H$ be Lie group homomorphisms so that the Lie algebra homomorphisms $df_1, df_2: \mathfrak{g} \rightarrow \mathfrak{h}$ are identical. Then $f_1 \equiv f_2$.

Proof. Since $df_1 = df_2$ as maps $\mathfrak{g} \rightarrow \mathfrak{h}$, their dual maps $f_1^* = f_2^*$ as maps $\mathfrak{h}^* \rightarrow \mathfrak{g}^*$, where we identify \mathfrak{g}^* and \mathfrak{h}^* with the left-invariant 1-forms on G and H , respectively.

Suppose $\omega_1, \dots, \omega_n$ is a basis for the left-invariant 1-forms on H . Then we know that $f_1^* \omega_i = f_2^* \omega_i$ for all $i = 1, \dots, n$. By uniqueness in [Theorem 3.9.1](#), it will follow that $f_1 \equiv f_2$ if we can show that the ideal \mathcal{I} on $G \times H$ generated by the

$$\mu_i = \pi_1^* f_1^* \omega_i - \pi_2^* \omega_i$$

is a differential ideal. It suffices to show that the $d\mu_i \in \mathcal{I}$, since a general element of \mathcal{I} will be a linear combination of terms of the form $\beta_1 \wedge \dots \wedge \beta_\ell \wedge \mu_i \wedge \beta_{\ell+1} \wedge \dots \wedge \beta_k$, the exterior derivative of which will be a linear combination of wedge products which always include either μ_i or $d\mu_i$.

So let's compute. First, notice that $\omega_1, \dots, \omega_n$ is a basis for the left-invariant forms on H , so $\{\omega_i \wedge \omega_j : i < j\}$ is a basis for the left-invariant 2-forms on H . Since $d\omega_k$ is left-invariant for any $k \in \{1, \dots, n\}$, there exist constants c_{ij}^k so that

$$d\omega_k = \sum_{i < j} c_{ij}^k \omega_i \wedge \omega_j.$$

But then

$$\begin{aligned} d\mu_k &= d(\pi_1^* f_1^* \omega_k - \pi_2^* \omega_k) \\ &= \pi_1^* f_1^* d\omega_k - \pi_2^* d\omega_k \\ &= \pi_1^* f_1^* \sum_{i < j} c_{ij}^k \omega_i \wedge \omega_j - \pi_2^* \sum_{i < j} c_{ij}^k \omega_i \wedge \omega_j \\ &= \sum_{i < j} c_{ij}^k (\pi_1^* f_1^* \omega_i \wedge \pi_1^* f_1^* \omega_j - \pi_2^* \omega_i \wedge \pi_2^* \omega_j) \\ &= \sum_{i < j} c_{ij}^k ((\pi_1^* f_1^* \omega_i - \pi_2^* \omega_i) \wedge \pi_1^* f_1^* \omega_j + \pi_2^* \omega_i \wedge (\pi_1^* f_1^* \omega_j - \pi_2^* \omega_j)) \\ &= \sum_{i < j} c_{ij}^k (\mu_i \wedge \pi_1^* f_1^* \omega_j + \pi_2^* \omega_i \wedge \mu_j) \end{aligned}$$

by adding and subtracting a term of the form $\pi_2^* \omega_i \wedge \pi_1^* f_1^* \omega_j$ in going from the fourth to the fifth line. But then each $\mu_i \wedge \pi_1^* f_1^* \omega_j + \pi_2^* \omega_i \wedge \mu_j \in \mathcal{I}$, so we conclude that $d\mu_k \in \mathcal{I}$, and hence \mathcal{I} is a differential ideal. \square

With all of this done, we have now proved uniqueness in [Theorem 3.6.1](#), which I restate:

Theorem (Theorem 3.6.1). Let G and H be Lie groups with Lie algebras \mathfrak{g} and \mathfrak{h} , respectively, and with G simply-connected. Let $\psi: \mathfrak{g} \rightarrow \mathfrak{h}$ be a Lie algebra homomorphism. Then there exists a unique Lie group homomorphism $\phi: G \rightarrow H$ so that $d\phi = \psi$.

I also restate [Corollary 3.6.2](#), which follows immediately:

Corollary (Corollary 3.6.2). If G and H are simply-connected and have isomorphic Lie algebras, then $G \cong H$.

Since we have proved uniqueness in Theorem 3.6.1, it remains to show existence, which we do by defining $\phi: G \rightarrow H$ as in the proof of Theorem 3.9.1 by $\phi = \pi_2 \circ (\pi_1|_S)^{-1}$ with $S \subset G \times H$ with $\mathcal{I} = \mathcal{I}(S)$. The key point is that the definition of the $\mu_i = \pi_1^* \psi^* \omega_i - \pi_2^* \omega_i$ which generate \mathcal{I} depends only on $\psi: \mathfrak{g} \rightarrow \mathfrak{h}$ and not on the ϕ that needs to be constructed. Proving that ϕ is really a homomorphism depends on some covering space theory that I don't really want to get into, but suffice it to say that this is true.

Notice that the simply-connected hypothesis in Corollary 3.6.2 is essential, since $O(n)$ and $SO(n)$ have identical Lie algebras even though they're not isomorphic as Lie groups (nor even homeomorphic). Admittedly, this feels like a cheat since it's really only using connectedness, and of course $O(n)$ and $SO(n)$ do have the same connected component of the identity, which is all that the Lie algebra could plausibly detect.

But we've already seen a less trivial example in Proposition 3.4.3, which showed that $\mathfrak{so}(3) \cong \mathfrak{su}(2)$, yet $SO(3)$ and $SU(2)$ are not isomorphic as groups (nor even homeomorphic as topological spaces). It is fairly easy to see that these groups are not isomorphic since they have different centers:

$$Z(SO(3)) = \{I\}, \quad \text{but} \quad Z(SU(2)) = \{I, -I\} \cong \mathbb{Z}/2\mathbb{Z}.$$

In fact, as topological spaces $SO(3) \cong \mathbb{RP}^3$ and $SU(2) \cong S^3$, so these spaces have different fundamental groups: $\pi_1(\mathbb{RP}^3) \cong \mathbb{Z}/2\mathbb{Z}$ and $\pi_1(S^3) = \{1\}$. Hence, they cannot be homeomorphic either. Looked at this way, S^3 is the universal cover of \mathbb{RP}^3 , meaning that $SU(2)$ is homeomorphic to the universal cover of $SO(3)$, and hence also to $\text{Spin}(3)$.⁵

3.10 The Exponential Map

In classifying Lie groups (especially compact Lie groups), it turns out to be particularly helpful to understand the abelian subgroups, and especially the maximal abelian subgroups (so-called *maximal tori*). The simplest abelian subgroups are called one-parameter subgroups:

Definition 3.10.1. A homomorphism $\phi: \mathbb{R} \rightarrow G$ with closed image (or sometimes the image of such a homomorphism) is called a *one-parameter subgroup* of G .

Example 3.10.2. The map $\phi: \mathbb{R} \rightarrow U(1) \times U(1)$ defined by $\phi(t) = (e^{it}, e^{i\sqrt{2}t})$ is *not* a one-parameter subgroup because its image is dense, and hence not closed.

Example 3.10.3. Define $\phi: \mathbb{R} \rightarrow SO(3)$ by

$$\phi(t) = \begin{bmatrix} \cos t & -\sin t & 0 \\ \sin t & \cos t & 0 \\ 0 & 0 & 1 \end{bmatrix}.$$

Then ϕ is a one-parameter subgroup of $SO(3)$, which corresponds to rotations around the z -axis. Of course, ϕ is periodic and in fact descends to an injective homomorphism $SO(2) \rightarrow SO(3)$ (where $SO(2)$ is diffeomorphic to the circle).

⁵In general, $\text{Spin}(n)$ is the universal cover of $SO(n)$.

Notice that

$$\phi'(0) = \begin{bmatrix} 0 & -1 & 0 \\ 1 & 0 & 0 \\ 0 & 0 & 0 \end{bmatrix} \in \mathfrak{so}(3);$$

in fact, this is one of the basis elements we found in the proof of [Proposition 3.4.3](#). Indeed, $\phi(t) = \gamma_3(t)$ from HW 3, Problem 2.

In that notation, we also have

$$\gamma_1(t) = \begin{bmatrix} 1 & 0 & 0 \\ 0 & \cos t & -\sin t \\ 0 & \sin t & \cos t \end{bmatrix} \quad \text{and} \quad \gamma_2(t) = \begin{bmatrix} \cos t & 0 & \sin t \\ 0 & 1 & 0 \\ -\sin t & 0 & \cos t \end{bmatrix},$$

and these also define one-parameter subgroups. More generally, if $a, b, c \in \mathbb{R}$, then

$$\gamma(t) := \gamma_1(at)\gamma_2(bt)\gamma_3(ct)$$

defines a one-parameter subgroup with

$$\gamma'(0) = a\gamma'_1(0)\gamma_2(0)\gamma_3(0) + b\gamma_1(0)\gamma'_2(0)\gamma_3(0) + c\gamma_1(0)\gamma_2(0)\gamma'_3(0) = \begin{bmatrix} 0 & -c & b \\ c & 0 & -a \\ -b & a & 0 \end{bmatrix},$$

so we see that every element of $\mathfrak{so}(3)$ determines a one-parameter subgroup.

This is a general phenomenon: if G is a Lie group and $X \in \mathfrak{g}$, then the map $\lambda \frac{d}{dr} \mapsto \lambda X$ defines a homomorphism from the Lie algebra of \mathbb{R} into \mathfrak{g} . Since \mathbb{R} is simply-connected, [Theorem 3.6.1](#) guarantees there exists a unique one-parameter subgroup $\phi_X : \mathbb{R} \rightarrow G$ so that $d\phi_X\left(\lambda \frac{d}{dr}\right) = \lambda X$.

Definition 3.10.4. Define the *exponential map* $\exp : \mathfrak{g} \rightarrow G$ by $\exp(X) = \phi_X(1)$.

Example 3.10.5. Let $G = \mathrm{U}(1)$, the unit complex numbers. We can identify $\mathfrak{u}(1)$ with the tangent space at the identity $T_1 \mathrm{U}(1)$, which is visually just a copy of the imaginary axis, as we see in [Figure 3.8](#).

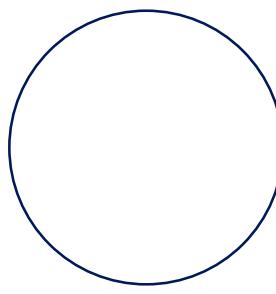


Figure 3.8: The tangent space at 1 to the unit circle.

We can make the identification $\mathfrak{u}(1) \cong i\mathbb{R}$. Let $X = ia \in \mathfrak{u}(1)$ and define the map $\phi_X : \mathbb{R} \rightarrow \mathrm{U}(1)$ by $\phi_X(r) = e^{rX} = e^{iar}$. Since $\phi'_X(0) = ia = X$, this is the one-parameter subgroup guaranteed by [Theorem 3.6.1](#), and so by definition

$$\exp(X) = \phi_X(1) = e^X = e^{ia}.$$

This helps explain the source of the name: the exponential map on $U(1)$ is just the usual exponential map from complex analysis.

More generally, suppose $A \in \text{Mat}_{n \times n}(\mathbb{R}) \cong \mathfrak{gl}_n(\mathbb{R})$ and define $\gamma(t) = e^{tA}$, where the matrix exponential is defined by the Taylor series

$$e^M := I + M + \frac{1}{2!}M^2 + \frac{1}{3!}M^3 + \dots$$

(which turns out to converge everywhere, just like the usual Taylor series for the single variable exponential). Then

$$\gamma'(0) = \left. \frac{d}{dt} \right|_{t=0} e^{tA} = Ae^{tA}|_{t=0} = A,$$

so $\gamma(t) = \phi_X(t)$ and hence $\exp(A) = \gamma(1) = e^A$. In other words, the exponential map on matrix groups is always just the matrix exponential.

Example 3.10.6. Let

$$A = \begin{bmatrix} 0 & -1 & 0 \\ 1 & 0 & 0 \\ 0 & 0 & 0 \end{bmatrix} \in \mathfrak{so}(3).$$

By uniqueness, we know that

$$e^{tA} = \exp(tA) = \begin{bmatrix} \cos t & -\sin t & 0 \\ \sin t & \cos t & 0 \\ 0 & 0 & 1 \end{bmatrix},$$

our one-parameter subgroup from [Example 3.10.3](#). But we can also prove this directly by computing powers of A :

$$A = \begin{bmatrix} 0 & -1 & 0 \\ 1 & 0 & 0 \\ 0 & 0 & 0 \end{bmatrix}, \quad A^2 = \begin{bmatrix} -1 & 0 & 0 \\ 0 & -1 & 0 \\ 0 & 0 & 0 \end{bmatrix}, \quad A^3 = \begin{bmatrix} 0 & 1 & 0 \\ -1 & 0 & 0 \\ 0 & 0 & 0 \end{bmatrix}, \quad A^4 = \begin{bmatrix} 1 & 0 & 0 \\ 0 & 1 & 0 \\ 0 & 0 & 0 \end{bmatrix}.$$

After this, the powers repeat, so we see that

$$e^{tA} = I + tA + \frac{1}{2!}(tA)^2 + \dots = \begin{bmatrix} 1 - \frac{t^2}{2!} + \frac{t^4}{4!} & -t + \frac{t^3}{3!} - \frac{t^5}{5!} & 0 \\ t - \frac{t^3}{3!} + \frac{t^5}{5!} & 1 - \frac{t^2}{2!} + \frac{t^4}{4!} & 0 \\ 0 & 0 & 0 \end{bmatrix} = \begin{bmatrix} \cos t & -\sin t & 0 \\ \sin t & \cos t & 0 \\ 0 & 0 & 0 \end{bmatrix}$$

using the Taylor series for sine and cosine.

Example 3.10.7. More generally, consider

$$X = \begin{bmatrix} 0 & -z & y \\ z & 0 & -x \\ -y & x & 0 \end{bmatrix} \in \mathfrak{so}(3),$$

where $x^2 + y^2 + z^2 = 1$. That is, $v = (x, y, z) \in \mathbb{R}^3$ is a unit vector and $X = F(v)$ in the notation of [Example 1.6.10](#).

The characteristic polynomial of X is

$$P(\lambda) = \det(\lambda I - X) = \det \begin{bmatrix} \lambda & z & -y \\ -z & \lambda & x \\ y & -x & \lambda \end{bmatrix} = \lambda^3 + \lambda(x^2 + y^2 + z^2) = \lambda^3 + \lambda,$$

so the Cayley–Hamilton theorem implies $0 = P(X) = X^3 + X$, so $X^3 = -X$, and hence

$$\begin{aligned} \exp(tX) &= I + tX + \frac{1}{2!}(tX)^2 + \frac{1}{3!}(tX)^3 + \dots \\ &= I + tX + \frac{t^2}{2!}X^2 - \frac{t^3}{3!}X - \frac{t^4}{4!}X^2 + \dots \\ &= I + \left(t - \frac{t^3}{3!} + \dots \right) X + \left(\frac{t^2}{2!} - \frac{t^4}{4!} + \dots \right) X^2 \\ &= I + (\sin t)X + (1 - \cos t)X^2. \end{aligned}$$

Therefore, for $u = (a, b, c) \in \mathbb{R}^3$, we have that

$$Xu = \begin{bmatrix} 0 & -z & y \\ z & 0 & -x \\ -y & x & 0 \end{bmatrix} \begin{bmatrix} a \\ b \\ c \end{bmatrix} = \begin{bmatrix} -bz + cy \\ az - cx \\ -ay + bx \end{bmatrix} = v \times u$$

and

$$X^2u = \begin{bmatrix} x^2 - 1 & xy & xz \\ xy & y^2 - 1 & yz \\ xz & yz & z^2 - 1 \end{bmatrix} \begin{bmatrix} a \\ b \\ c \end{bmatrix} = \begin{bmatrix} a(x^2 - 1) + bxy + cxz \\ axy + b(y^2 - 1) + cyz \\ axz + byz + c(z^2 - 1) \end{bmatrix} = (u \cdot v)v - u.$$

Therefore,

$$\begin{aligned} \exp(tX)u &= (I + (\sin t)X + (1 - \cos t)X^2)u = u + (\sin t)v \times u + (1 - \cos t)((u \cdot v)v - u) \\ &= (\cos t)u + (\sin t)v \times u + (1 - \cos t)(u \cdot v)v, \end{aligned}$$

which is just the Rodrigues formula for rotation of u by an angle t around the axis determined by v .

So this tells us that the one-parameter subgroups of $\mathrm{SO}(3)$ correspond to rotations around axes, where, if $X = F(v)$, then v is the axis of rotation. Notice that the map $t \mapsto \exp(tX)$ is 2π -periodic so, while in principle this 1-parameter subgroup is just a homomorphism $\mathbb{R} \rightarrow \mathrm{SO}(3)$, in fact it descends to a homomorphism $S^1 \rightarrow \mathrm{SO}(3)$. If we interpret the circle group S^1 as $\mathrm{SO}(2)$, then this is just a natural embedding of $\mathrm{SO}(2)$ into $\mathrm{SO}(3)$ by choosing an axis to rotate around.

Example 3.10.8. Recall from the proof of Proposition 3.4.3 that $\mathfrak{su}(2)$ consists of traceless, skew-Hermitian 2×2 matrices, so I can represent an arbitrary element of $\mathfrak{su}(2)$ as $X = \begin{bmatrix} ix & y + iz \\ -y + iz & -ix \end{bmatrix}$ for $x, y, z \in \mathbb{R}$. If I assume $x^2 + y^2 + z^2 = 1$, then the characteristic polynomial is

$$P(\lambda) = \det(\lambda I - X) = \det \begin{bmatrix} \lambda - ix & -y - iz \\ y - iz & \lambda - ix \end{bmatrix} = \lambda^2 + x^2 + y^2 + z^2 = \lambda^2 + 1,$$

so Cayley–Hamilton implies $0 = P(X) = X^2 + I$, so $X^2 = -I$. Therefore,

$$\begin{aligned}\exp(tX) &= I + tX + \frac{1}{2!}(tX)^2 + \dots = I + tX - \frac{t^2}{2!}I - \frac{t^3}{3!}X + \dots \\ &= \left(1 - \frac{t^2}{2!} + \dots\right)I + \left(t - \frac{t^3}{3!} + \dots\right)X = (\cos t)I + (\sin t)X = \begin{bmatrix} \cos t + ix \sin t & (y + iz) \sin t \\ (y - iz) \sin t & \cos t - ix \sin t \end{bmatrix}.\end{aligned}$$

As in the previous example, this map is 2π -periodic, so the one-parameter subgroup descends to a homomorphism $S^1 \rightarrow \mathrm{SU}(2)$. In this setting, it's more natural to identify the circle with $\mathrm{U}(1)$.

Some straightforward properties of the exponential:

Proposition 3.10.9. *Let $X \in \mathfrak{g}$. Then*

- (i) $\exp((t_1 + t_2)X) = \exp(t_1 X)\exp(t_2 X)$ for all $t_1, t_2 \in \mathbb{R}$
- (ii) $\exp(-tX) = (\exp(tX))^{-1}$ for all $t \in \mathbb{R}$
- (iii) $\exp: \mathfrak{g} \rightarrow G$ is smooth and $(d\exp)_0: T_0\mathfrak{g} \rightarrow T_eG$ is the identity map under the identification of $T_0\mathfrak{g}$ and T_eG with \mathfrak{g} . Hence, the inverse function theorem implies that \exp defines a diffeomorphism from a neighborhood of the origin in \mathfrak{g} to a neighborhood of the identity in G .

In general, the exponential map is how we translate between Lie algebra homomorphisms and Lie group homomorphisms:

Theorem 3.10.10. *Let $\phi: H \rightarrow G$ be a Lie group homomorphism. Then the following diagram commutes:*

$$\begin{array}{ccc} H & \xrightarrow{\phi} & G \\ \exp \uparrow & & \uparrow \exp \\ \mathfrak{h} & \xrightarrow{d\phi} & \mathfrak{g} \end{array}$$

Proof. Let $X \in \mathfrak{h}$. Then $\gamma(t) := \phi(\exp(tX))$ is certainly a smooth curve in G with

$$\gamma'(0) = \left. \frac{d}{dt} \right|_{t=0} (\phi \circ \exp)(tX) = d\phi(X).$$

Moreover, it's a one-parameter subgroup since G is a homomorphism, so by uniqueness

$$\gamma(t) = \exp(td\phi(X))$$

for all t , including $t = 1$, which gives $\phi(\exp(X)) = \gamma(1) = \exp(d\phi(X))$, as desired. \square

Example 3.10.11. As in [Example 3.10.8](#), let $X = \begin{bmatrix} ix & y + iz \\ -y + iz & -ix \end{bmatrix} \in \mathfrak{su}(2)$. As we essentially showed in the proof of [Proposition 3.4.3](#), the map $\psi: \mathfrak{su}(2) \rightarrow \mathfrak{so}(3)$ defined by

$$\psi \left(\begin{bmatrix} ix & y + iz \\ -y + iz & -ix \end{bmatrix} \right) := \begin{bmatrix} 0 & -2z & 2y \\ 2z & 0 & -2x \\ -2y & 2x & 0 \end{bmatrix}$$

is a Lie algebra homomorphism (in fact, isomorphism). Since $\mathrm{SU}(2)$ is simply-connected, [Theorem 3.6.1](#) implies that there exists a Lie group homomorphism $\phi: \mathrm{SU}(2) \rightarrow \mathrm{SO}(3)$ so that $\psi = d\phi$.

By [Theorem 3.10.10](#), we know that

$$\phi(\exp(X)) = \exp(d\phi(X)) = \exp(\psi(X)).$$

Since the image of ϕ is contained in $\mathrm{SO}(3)$, $\phi(\exp(X))$ acts on vectors $v \in \mathbb{R}^3$. By combining with [Example 3.10.7](#), we can see that $\phi(\exp(X))$ acts by rotating around the axis $v = (x, y, z)$ at double speed.

In particular, if we assume $\|v\| = 1$, then $\exp\left(\frac{\pi}{2}X\right) \neq \exp\left(-\frac{\pi}{2}X\right)$ as elements of $\mathrm{SU}(2)$, but

$$\phi\left(\exp\left(\frac{\pi}{2}X\right)\right) = \phi\left(\exp\left(-\frac{\pi}{2}X\right)\right)$$

as elements of $\mathrm{SO}(3)$ since the first rotates around axis v by π radians, and the second by $-\pi$ radians, which of course is the same rotation.

This reflects the fact that the map $\phi: \mathrm{SU}(2) \rightarrow \mathrm{SO}(3)$ is a double-covering.

3.11 Homogeneous Manifolds

Given a group G and a normal subgroup N , it's a standard fact from algebra that the quotient (i.e., collection of cosets) G/N is also a group. The key point is that, for $g_1, g_2 \in G$, the coset $(g_1N)(g_2N) := g_1g_2N$ is well-defined: for $n_1, n_2 \in N$, we can write $n_1 = g_2ng_2^{-1}$ for $n := g_2^{-1}n_1g_2 \in N$, so

$$(g_1n_1)(g_2n_2) = (g_1g_2ng_2^{-1})(g_2n_2) = (g_1g_2)(nn_2) \in g_1g_2N.$$

The same holds true for Lie groups: if G is a Lie group and N is a normal Lie subgroup, then G/N is a Lie group. For example, $\mathrm{SL}_n(\mathbb{R})$ is a closed normal subgroup of $\mathrm{GL}_n(\mathbb{R})$, and

$$\mathrm{GL}_n(\mathbb{R})/\mathrm{SL}_n(\mathbb{R}) \cong \mathbb{R}^\times = \mathrm{GL}_1(\mathbb{R}),$$

the multiplicative group of nonzero real numbers

Okay, but what if we take a Lie subgroup H of a Lie group G which is not normal? Does the coset space G/H have any special structure? Not surprisingly, given how I've set it up, the answer is yes:

Theorem 3.11.1. *Let H be a closed subgroup of a Lie group G (i.e., a Lie subgroup) and let G/H be the set of left cosets $\{gH : g \in G\}$. If $\pi: G \rightarrow G/H$ is the obvious map, then G/H has a unique manifold structure so that*

- (i) π is smooth and
- (ii) there exist local sections of G/H : for $gH \in G/H$, there exists a neighborhood U of gH and a smooth map $\sigma: U \rightarrow G$ so that $\pi \circ \sigma = \mathrm{id}|_U$.

The proof of this theorem is kind of a pain, so we skip it. Manifolds of the form G/H are called *homogeneous manifolds* (or sometimes *homogeneous spaces*).

Example 3.11.2. Let $G = \text{SO}(3)$ and let $H = \left\{ \begin{bmatrix} \cos \theta & -\sin \theta & 0 \\ \sin \theta & \cos \theta & 0 \\ 0 & 0 & 1 \end{bmatrix} : \theta \in \mathbb{R} \right\}$. Then the theorem says that G/H is a manifold.

Which one? Well, think about the usual action of $\text{SO}(3)$ on \mathbb{R}^3 (or equivalently, by [Example 3.5.5](#), the adjoint representation of $\text{SO}(3)$). If we pick $v = (0, 0, 1) \in \mathbb{R}^3$, then the orbit $G \cdot v = \{g \cdot v : g \in \text{SO}(3)\}$ (i.e., the adjoint orbit of the corresponding element of $\mathfrak{so}(3)$) will consist of all vectors with the same norm as v ; in other words, it is the standard sphere of radius 1. Moreover, the action of H on v is trivial: $H \cdot v = \{v\}$. So it is not insane to guess that the elements of G/H are in one-to-one correspondence with the points on the sphere.

Indeed, this is true: define the map $\phi: \text{SO}(3)/H \rightarrow S^2$ by

$$gH \mapsto g \cdot v.$$

I claim this is a diffeomorphism, so there are a bunch of things to check:

ϕ is well-defined: For $h \in H$, $\phi(ghH) = (gh) \cdot v = g \cdot (h \cdot v) = g \cdot v = \phi(gH)$ since $h \cdot v = v$.

ϕ is surjective: It's geometrically clear that the action of $\text{SO}(3)$ on S^2 is transitive, but let's actually see

this explicitly. Suppose $u \in S^2$. If $u = (0, 0, \pm 1)$ and $g = \begin{bmatrix} \pm 1 & 0 & 0 \\ 0 & 1 & 0 \\ 0 & 0 & \pm 1 \end{bmatrix}$, then $g \cdot v = u$. Otherwise,

$u = (x, y, z)$ where $x^2 + y^2 > 0$. If θ is the angle between u and v , then $\cos \theta = u \cdot v = z$ and $x^2 + y^2 = 1 - z^2 = 1 - \cos^2 \theta = \sin^2 \theta$. Define $n := \frac{u \times v}{\|u \times v\|} = \frac{(y, -x, 0)}{\sqrt{x^2 + y^2}}$, and let N be the corresponding element of $\mathfrak{so}(3)$:

$$N = \begin{bmatrix} 0 & 0 & \frac{-x}{\sqrt{x^2+y^2}} \\ 0 & 0 & \frac{-y}{\sqrt{x^2+y^2}} \\ \frac{x}{\sqrt{x^2+y^2}} & \frac{y}{\sqrt{x^2+y^2}} & 0 \end{bmatrix}.$$

Then I claim that $e^{-\theta N} v = u$. To see this, we can use [Example 3.10.7](#) to compute

$$\begin{aligned} e^{-\theta N} v &= \cos(-\theta)v + \sin(-\theta)n \times v + (1 - \cos(-\theta))(v \cdot n)n \\ &= (\cos \theta) \begin{bmatrix} 0 \\ 0 \\ 1 \end{bmatrix} - \frac{\sin \theta}{\sqrt{x^2 + y^2}} \begin{bmatrix} -x \\ -y \\ \cos \theta - z \end{bmatrix} + 0 = \begin{bmatrix} x \\ y \\ z \end{bmatrix} = u \end{aligned}$$

since $n \times v = (u \times v) \times v = (u \cdot v)v - (v \cdot v)u = (\cos \theta)v - u$ and $\sqrt{x^2 + y^2} = \sin \theta$.

ϕ is injective: Suppose $\phi(g_1H) = \phi(g_2H)$. Then

$$\begin{aligned} \phi(g_2^{-1}g_1H) &= (g_2^{-1}g_1) \cdot v = g_2^{-1} \cdot (g_1 \cdot v) = g_2^{-1} \cdot \phi(g_1H) = g_2^{-1} \cdot \phi(g_2H) = g_2^{-1} \cdot (g_2 \cdot v) = (g_2^{-1}g_2) \cdot v = v \\ \text{so } g_2^{-1}g_1 &\in H \text{ (since } H \text{ consists of all elements of } \text{SO}(3) \text{ fixing } v\text{), and hence } g_2H = g_2g_2^{-1}g_1H = g_1H. \end{aligned}$$

ϕ is smooth: If $\pi: \mathrm{SO}(3) \rightarrow \mathrm{SO}(3)/H$ is the projection, then I claim that ϕ is smooth if and only if $\phi \circ \pi$ is smooth.

$$\begin{array}{ccc} \mathrm{SO}(3) & & \\ \pi \downarrow & \searrow^{\phi \circ \pi} & \\ \mathrm{SO}(3)/H & \xrightarrow{\phi} & S^2 \end{array}$$

If ϕ is smooth then $\phi \circ \pi$ is clearly smooth since π is. On the other hand, if $\phi \circ \pi$ is smooth, then on a neighborhood U of $gH \in \mathrm{SO}(3)/H$ with local section $\sigma: U \rightarrow \mathrm{SO}(3)$ we have that

$$(\phi \circ \pi) \circ \sigma = \phi \circ (\pi \circ \sigma) = \phi \circ \mathrm{id}|_U = \phi|_U$$

is smooth. Since such local sections exist at all points (by [Theorem 3.11.1](#)), ϕ is smooth. But now it's clear that $(\phi \circ \pi)(g) = g \cdot v$ is smooth since the rotation action on S^2 is smooth, and therefore ϕ is smooth.

ϕ has smooth inverse: We know ϕ is bijective, so it has an inverse, and the only challenge is to show that the inverse is smooth. By the inverse function theorem, it suffices to show that $d\phi$ is nonsingular everywhere, or equivalently that $\ker d(\phi \circ \pi)_g \subset T_g H \subset T_g \mathrm{SO}(3)$. Now, for $g \in \mathrm{SO}(3)$,

$$\begin{aligned} (\phi \circ \pi)(g) &= \phi(\pi(g)) = \phi(gH) = g \cdot v = g \cdot (I \cdot v) = g \cdot (\phi(H)) = g \cdot ((\phi \circ \pi)(I)) \\ &= g \cdot ((\phi \circ \pi)(g^{-1}g)) = g \cdot ((\phi \circ \pi \circ L_{g^{-1}})(g)), \end{aligned}$$

so it suffices to show this at the identity: $\ker d(\phi \circ \pi)_I \subset T_I H$. So now let $U = \begin{bmatrix} 0 & -z & y \\ z & 0 & -x \\ -y & x & 0 \end{bmatrix} \in T_I \mathrm{SO}(3)$, corresponding to the vector $u = (x, y, z) \in S^2$. We know that $U = \alpha'(0)$ where $\alpha(t) = e^{tU}$, so

$$\begin{aligned} d(\phi \circ \pi)_I(U) &= ((\phi \circ \pi) \circ \alpha)'(0) = \frac{d}{dt} \Big|_{t=0} \phi(\pi(\alpha(t))) \\ &= \frac{d}{dt} \Big|_{t=0} \phi(\pi(e^{tU})) = \frac{d}{dt} \Big|_{t=0} \phi(e^{tU} H) = \frac{d}{dt} \Big|_{t=0} e^{tU} \cdot v. \end{aligned}$$

Again, we can use [Example 3.10.7](#) to compute

$$e^{tU} \cdot v = (\cos t)v + (\sin t)u \times v + (1 - \cos t)(v \cdot u)u = \begin{bmatrix} y \sin t + xz(1 - \cos t) \\ -x \sin t + yz(1 - \cos t) \\ z \cos t + z^2(1 - \cos t) \end{bmatrix},$$

so

$$d(\phi \circ \pi)_I(U) = \frac{d}{dt} \Big|_{t=0} \begin{bmatrix} y \sin t + xz(1 - \cos t) \\ -x \sin t + yz(1 - \cos t) \\ z \cos t + z^2(1 - \cos t) \end{bmatrix} = \begin{bmatrix} y \\ -x \\ 0 \end{bmatrix}.$$

So $U \in \ker d(\phi \circ \pi)_I$ if and only if $x = y = 0$, meaning that $U = \begin{bmatrix} 0 & -z & 0 \\ z & 0 & 0 \\ 0 & 0 & 0 \end{bmatrix}$, which is indeed in the tangent space to H .

Thus, we can conclude that $d\phi$ is nonsingular everywhere and hence the inverse is smooth.

Essentially the same argument gives a much more general result, which we need one definition in order to state:

Definition 3.11.3. If M is a manifold and G is a Lie group which acts smoothly on M , the *isotropy subgroup* of G at a point $p \in M$ is the group

$$G_p = \{g \in G : g \cdot p = p\}.$$

(Incidentally, this induces the *isotropy representation* $G_p \rightarrow \text{Aut}(T_p M)$ given by $g \mapsto (d\phi_g)_p|_{T_p M}$, where $\phi_g(q) := g \cdot q$ for all $q \in M$.)

Theorem 3.11.4. Suppose a Lie group G acts transitively on a manifold M and that, for some $p \in M$, $H = G_p$ is the isotropy subgroup for p . Then M is diffeomorphic to G/H .

Indeed, most books define *homogeneous manifolds* to be manifolds admitting a smooth, transitive action of a Lie group, which explains the name: since the group action is transitive, every point is in some sense the same as every other point. [Theorem 3.11.4](#) tells us that our definition of homogeneous manifold and the standard definition are equivalent.

Example 3.11.5. The same reasoning as in [Example 3.11.2](#) shows that $S^n \cong \text{SO}(n+1)/\text{SO}(n)$.

Example 3.11.6. Consider $\text{U}(n+1)$ acting on the unit sphere $S^{2n+1} \subset \mathbb{C}^{n+1}$. The isotropy subgroup of the north pole is the subgroup

$$\left\{ \begin{bmatrix} U & 0 \\ 0 & 1 \end{bmatrix} : U \in \text{U}(n) \right\},$$

so $S^{2n+1} \cong \text{U}(n+1)/\text{U}(n)$.

Consider the special case when $n = 0$. Then this says $S^1 \cong \text{U}(1)/\text{U}(0)$, which is obviously true since $\text{U}(0) = \{1\}$ and $\text{U}(1)$ is just the unit complex numbers; i.e., the unit circle.

Example 3.11.7. Similarly, $S^{2n+1} \cong \text{SU}(n+1)/\text{SU}(n)$. In the special case $n = 1$, we get

$$S^3 \cong \text{SU}(2)/\text{SU}(1) = \text{SU}(2)/\{1\} \cong \text{SU}(2),$$

which gives a simple proof of a fact we've alluded to many times.

3.12 Stiefel and Grassmann Manifolds

Stiefel manifolds and Grassmann manifolds (or Grassmannians) are very important classes of homogeneous manifolds which show up all over the place in both pure and applied mathematics. In turn, the simplest example of a Grassmannian is a projective space:

Example 3.12.1. Complex projective space \mathbb{CP}^n is the collection of all 1-dimensional subspaces of \mathbb{C}^{n+1} ; equivalently,

$$\mathbb{CP}^n = \{(x_0, \dots, x_n) \in \mathbb{C}^{n+1} - \{0\}\} / \sim,$$

where $(x_0, \dots, x_n) \sim \lambda(x_0, \dots, x_n)$ for all $\lambda \in \mathbb{C}^\times$.

If we restrict to unit vectors, we can also see

$$\mathbb{CP}^n = \{(x_0, \dots, x_n) \in S^{2n+1} \subset \mathbb{C}^{n+1}\} / \sim,$$

where now $(x_0, \dots, x_n) \sim \lambda(x_0, \dots, x_n)$ for all $\lambda \in U(1)$.

We typically represent the equivalence class of a point (x_0, \dots, x_n) by *homogeneous coordinates* $[x_0 : \dots : x_n]$.

As we saw in [Example 3.11.7](#), $SU(n+1)$ acts transitively on S^{2n+1} . Moreover, this action preserves equivalence classes since

$$A(\lambda(x_0, \dots, x_n)) = \lambda A(x_0, \dots, x_n) \sim A(x_0, \dots, x_n)$$

by complex-linearity of A , so $SU(n+1)$ acts transitively on \mathbb{CP}^n . The isotropy subgroup of $[1 : 0 : \dots : 0] \in \mathbb{CP}^n$ is just

$$\left\{ \begin{bmatrix} \frac{1}{\det A} & 0 \\ 0 & A \end{bmatrix} : A \in U(n) \right\} \cong U(n),$$

so $\mathbb{CP}^n \cong SU(n+1)/U(n)$ is a homogeneous space.

Moreover, since $SU(n+1) \cong U(n)/U(1)$ we have that

$$\begin{aligned} \mathbb{CP}^n &\cong SU(n+1)/U(n) \cong (U(n+1)/U(1))/U(n) \cong U(n+1)/(U(1) \times U(n)) \\ &\cong (U(n+1)/U(n))/U(1) \cong S^{2n+1}/U(1) \end{aligned}$$

by [Example 3.11.6](#). Therefore, we have a fibration

$$\begin{array}{ccc} U(1) & \hookrightarrow & S^{2n+1} \\ & \searrow & \\ & & \mathbb{CP}^n \end{array}$$

Another interpretation: the intersection of each complex line with S^{2n+1} is a circle of the form λv where λ is a unit complex number and $v \in S^{2n+1}$. So then the equivalence classes of such circles exactly correspond to complex lines in \mathbb{C}^{n+1} .

Notice that the fiber is a sphere: $U(1) \cong S^1$.

In the special case $n = 1$, the resulting fibration is

$$\begin{array}{ccc} U(1) & \hookrightarrow & S^3 \\ & \searrow & \\ & & \mathbb{CP}^1 \cong S^2 \end{array}$$

usually called the *Hopf fibration*; here's a nice [Hopf fibration video by Niles Johnson](#).

Based on this, the fibrations for $n > 1$ are sometimes called *generalized Hopf fibrations*. We can play similar games with \mathbb{HP}^n and \mathbb{OP}^1 to get other sphere fibrations.

While the above is an example of a Grassmannian, let's define Stiefel manifolds before defining Grassmannians:

Definition 3.12.2. The *Stiefel manifold* $St(k, \mathbb{R}^n)$ of (orthonormal) k -frames in \mathbb{R}^n is the collection

$$St(k, \mathbb{R}^n) := \{(v_1, \dots, v_k) : v_i \in \mathbb{R}^n \text{ and } \langle v_i, v_j \rangle = \delta_{ij} \text{ for all } 1 \leq i, j \leq k\}.$$

In other words, elements of $\text{St}(k, \mathbb{R}^n)$ are k -tuples of pairwise orthonormal vectors in \mathbb{R}^n (with respect to, say, the standard inner product). Of course this implies that v_1, \dots, v_k are linearly independent, so $k \leq n$.

Remark 3.12.3. The notations $V_k(\mathbb{R}^n)$, $V(k, n)$, and $V_{n,k}$ are often used instead of my $\text{St}(k, \mathbb{R}^n)$.

Example 3.12.4. If $k = 1$, then $\text{St}(1, \mathbb{R}^n) = S^{n-1}$, the unit sphere in \mathbb{R}^n .

Example 3.12.5. If $k = n$, then we can think of $(v_1, \dots, v_n) \in \text{St}(n, \mathbb{R}^n)$ as giving the columns of an orthogonal matrix, so $\text{St}(n, \mathbb{R}^n) \cong \text{O}(n)$ (as manifolds).

Remark 3.12.6. Suppose $(v_1, \dots, v_k) \in \text{St}(k, \mathbb{R}^n)$. Let V be the $n \times k$ matrix whose i th column is v_i . Then the condition $\langle v_i, v_j \rangle = \delta_{ij}$ can also be written as

$$V^T V = I,$$

the $d \times d$ identity matrix. In the language of frame theory, this means that the columns of V^T (or the rows of V) form a *Parseval frame*. In other words, Stiefel manifolds are precisely the spaces of (finite) Parseval frames of given dimensions.

$\text{O}(n)$ acts transitively on $\text{St}(k, \mathbb{R}^n)$, which we can see because any orthogonal matrix with (v_1, \dots, v_k) as its first k columns sends the first k standard basis vectors (e_1, \dots, e_k) (which is a particular point in the Stiefel manifold) to (v_1, \dots, v_k) . Therefore, $\text{St}(k, \mathbb{R}^n) \cong \text{O}(n)/H$ where $H \subset \text{O}(n)$ is an isotropy subgroup. Consider the point $(e_1, \dots, e_k) \in \text{St}(k, \mathbb{R}^n)$. Then the isotropy subgroup of this point is

$$\left\{ \begin{bmatrix} I_k & 0 \\ 0 & A \end{bmatrix} : A \in \text{O}(n-k) \right\} \cong \text{O}(n-k),$$

so $\text{St}(k, \mathbb{R}^n) \cong \text{O}(n)/\text{O}(n-k)$. Notice that this agrees with $S^{n-1} \cong \text{O}(n)/\text{O}(n-1)$ when $k = 1$ and with $\text{O}(n) \cong \text{O}(n)/\text{O}(0) = \text{O}(n)/\{I\}$ when $k = n$.

In particular, this implies that

$$\dim(\text{St}(k, \mathbb{R}^n)) = \dim(\text{O}(n)) - \dim(\text{O}(n-k)) = \frac{n(n+1)}{2} - \frac{(n-k)(n-k+1)}{2} = nk - \frac{k(k-1)}{2}.$$

Definition 3.12.7. The *Grassmannian* (or *Grassmann manifold*) $\text{Gr}(k, \mathbb{R}^n)$ of k -planes in \mathbb{R}^n is the collection of k -dimensional linear subspaces of \mathbb{R}^n .

Remark 3.12.8. The notations $G_k(\mathbb{R}^n)$, $\text{Gr}(k, n)$, and $G_{k,n}$ are sometimes used for the Grassmannian.

Example 3.12.9. When $k = 1$, we're talking about 1-dimensional subspaces of \mathbb{R}^n ; in this case, the Grassmannian $\text{Gr}(1, \mathbb{R}^n)$ is usually called the *real projective space* \mathbb{RP}^{n-1} .

As in the case of the Stiefel manifold, $\text{O}(n)$ acts transitively on $\text{Gr}(k, \mathbb{R}^n)$, and the isotropy subgroup of the k -plane $\text{span}\{e_1, \dots, e_k\}$ is

$$\left\{ \begin{bmatrix} A & 0 \\ 0 & B \end{bmatrix} : A \in \text{O}(k), B \in \text{O}(n-k) \right\} \cong \text{O}(k) \times \text{O}(n-k),$$

so $\text{Gr}(k, \mathbb{R}^n) \cong \text{O}(n)/(\text{O}(k) \times \text{O}(n-k))$.

In particular, this shows that

$$\dim(\text{Gr}(k, \mathbb{R}^n)) = \dim(\text{O}(n)) - \dim(\text{O}(k)) - \dim(\text{O}(n-k)) = \frac{n(n+1)}{2} - \frac{k(k+1)}{2} - \frac{(n-k)(n-k+1)}{2} = k(n-k).$$

Taking quotients in stages reveals that

$$\mathrm{Gr}(k, \mathbb{R}^n) \cong \mathrm{O}(n)/(\mathrm{O}(k) \times \mathrm{O}(n-k)) \cong (\mathrm{O}(n)/\mathrm{O}(n-k))/\mathrm{O}(k) \cong \mathrm{St}(k, \mathbb{R}^n)/\mathrm{O}(k).$$

Here's a more geometric interpretation of this: given $P \in \mathrm{Gr}(k, \mathbb{R}^n)$, one could find an orthonormal basis v_1, \dots, v_k for $P = \mathrm{span}\{v_1, \dots, v_k\}$; then $(v_1, \dots, v_k) \in \mathrm{St}(k, \mathbb{R}^n)$ and the map $(v_1, \dots, v_k) \mapsto \mathrm{span}\{v_1, \dots, v_k\}$ defines a map $\mathrm{St}(k, \mathbb{R}^n) \rightarrow \mathrm{Gr}(k, \mathbb{R}^n)$. Two elements of the Stiefel manifold get sent to the same element of the Grassmannian if they are two different orthonormal bases for the same k -plane, which means they must be related by an orthogonal transformation of the k -plane. Identifying (v_1, \dots, v_k) with the (tall, skinny) matrix whose columns are the v_i , this corresponds to right-multiplying by an element of $\mathrm{O}(k)$, so indeed elements of $\mathrm{Gr}(k, \mathbb{R}^n)$ correspond to (right) $\mathrm{O}(k)$ -orbits of points in $\mathrm{St}(k, \mathbb{R}^n)$.

This perspective naturally leads to the so-called *Stiefel bundles* (or, in fancy language, the frame bundle associated to the tautological bundle of the Grassmannian):

$$\begin{array}{ccc} \mathrm{O}(k) & \hookrightarrow & \mathrm{St}(k, \mathbb{R}^n) \\ & & \downarrow \\ & & \mathrm{Gr}(k, \mathbb{R}^n) \end{array}$$

In general, if V is a vector space, we can define the Grassmannian $\mathrm{Gr}(k, V)$ to be the space of k -dimensional linear subspaces of V . Also, when V is an inner product space (e.g., \mathbb{R}^n or \mathbb{C}^n), we can define the Stiefel manifold $\mathrm{St}(k, V)$ of k -tuples of pairwise orthonormal vectors in V .

When $V = \mathbb{C}^n$, it is straightforward to show that $\mathrm{St}(k, \mathbb{C}^n) \cong \mathrm{U}(n)/\mathrm{U}(n-k)$ and $\mathrm{Gr}(k, \mathbb{C}^n) \cong \mathrm{U}(n)/(\mathrm{U}(k) \times \mathrm{U}(n-k))$. We get a corresponding complex Stiefel bundle:

$$\begin{array}{ccc} \mathrm{U}(k) & \hookrightarrow & \mathrm{St}(k, \mathbb{C}^n) \\ & & \downarrow \\ & & \mathrm{Gr}(k, \mathbb{C}^n) \end{array}$$

When $k = 1$ this is just the generalized Hopf fibration from [Example 3.12.1](#), so these Stiefel bundles are in some sense generalized generalized Hopf fibrations.

Remark 3.12.10. Returning to the real case (though something analogous is true in \mathbb{C}^n), there is a diffeomorphism $\mathrm{Gr}(k, \mathbb{R}^n) \rightarrow \mathrm{Gr}(n-k, \mathbb{R}^n)$ which maps a k -dimensional subspace $P \in \mathrm{Gr}(k, \mathbb{R}^n)$ to its orthogonal complement $P^\perp \in \mathrm{Gr}(n-k, \mathbb{R}^n)$. Combined with [Example 3.12.9](#), this shows that

$$\mathrm{Gr}(n-1, \mathbb{R}^n) \cong \mathrm{Gr}(1, \mathbb{R}^n) \cong \mathbb{RP}^{n-1}.$$

So the first Grassmannian which is not a projective space is $\mathrm{Gr}(2, \mathbb{R}^4)$, which has dimension $2(4-2) = 4$.

I will also briefly mention that Grassmannians are special cases of more general homogeneous manifolds called *flag manifolds* (or *flag varieties*). In short, for $1 \leq d_1 < d_2 < \dots < d_{k-1} < d_k = n$, a *flag of signature* (d_1, \dots, d_k) is a nested collection of subspaces of \mathbb{R}^n (or \mathbb{C}^n)

$$V_1 \subset V_2 \subset \dots \subset V_{k-1} \subset V_k = \mathbb{R}^n.$$

The collection of all flags of signature (d_1, \dots, d_k) is the flag manifold $\mathrm{Fl}(d_1, \dots, d_k)$. Notice that a flag of signature (d, n) is just a single d -dimensional subspace $V \subset \mathbb{R}^n$; that is, a point in the Grassmannian $\mathrm{Gr}(d, \mathbb{R}^n)$. Therefore, $\mathrm{Fl}(d, n) = \mathrm{Gr}(d, \mathbb{R}^n)$, so the flag manifolds really are generalizations of Grassmannians.

At the other end of the spectrum, flags of signature $(1, 2, \dots, n-1, n)$ are called *complete flags*, and $\mathrm{Fl}(1, 2, \dots, n)$ is called the *complete flag manifold* (or *complete flag variety*).⁶

Exercise 3.12.11. Show that

$$\mathrm{Fl}(d_1, \dots, d_k) \cong \mathrm{O}(n)/(\mathrm{O}(n_1) \times \cdots \times \mathrm{O}(n_k)),$$

where $n_1 = d_1$ and $n_i = d_i - d_{i-1}$ for $i = 2, \dots, k$. (In other words, the n_i are the jumps in dimension as we go up the flag.)

Remark 3.12.12. For this and other reasons, it is sometimes more convenient to think in terms of the n_i rather than the d_i and to talk about flags of type (n_1, n_2, \dots, n_k) and flag manifolds $\mathrm{Fl}(n_1, n_2, \dots, n_k)$. In this notation, the complete flag manifold is $\mathrm{Fl}(1, 1, \dots, 1)$ and the Grassmannian $\mathrm{Gr}(d, \mathbb{R}^n)$ is $\mathrm{Fl}(d, n-d)$.

The coadjoint orbits of the unitary group $\mathrm{U}(n)$ turn out to give all of the complex flag manifolds consisting of flags in \mathbb{C}^n . For more on this, see, e.g., Audin's book [2, §II.1.d].

⁶By contrast, flags which are not complete are sometimes—especially in the algebraic geometry literature—called *incomplete flags* and the associated manifolds *incomplete flag manifolds* (or *incomplete flag varieties*).

Chapter 4

Riemannian Manifolds

4.1 Riemannian Metrics

Thus far, almost all of what we've talked about is, strictly speaking, in the realm of *differential topology* rather than differential geometry. The exceptions include [Section 2.2](#), [Example 2.3.22](#), and some of the distribution theory (including contact geometry) from [Section 3.7](#) and [Section 3.8](#). The point is that, to have geometry we need some notion of size, whether that's in the form of length, area, volume, or something else.

A weak form of this is provided by volume forms, and a somewhat stronger form by symplectic forms, which encode a notion of area (and hence volume), but not of length. In this section we introduce Riemannian metrics, which induce a notion of length (and hence also of area, volume, etc.). We've already given a definition in [Definition 2.3.21](#), which we repeat:

Definition ([Definition 2.3.21](#)). A *Riemannian metric* on a manifold M is a smooth $(0, 2)$ -tensor field g on M so that, for each $p \in M$, $g_p = g(p)$ satisfies the axioms of an inner product on $T_p(M)$; that is, in addition to being bilinear (which is guaranteed by the fact it is a $(0, 2)$ -tensor), it must be symmetric ($g_p(u, v) = g_p(v, u)$) and positive-definite ($g_p(v, v) \geq 0$ with equality if and only if $v = 0$).

Definition 4.1.1. A pair (M, g) where M is a manifold and g is a Riemannian metric is called a *Riemannian manifold*.

Let (U, ϕ) be some local coordinate chart on M and let $\frac{\partial}{\partial x_1}, \dots, \frac{\partial}{\partial x_n}$ be the induced local coordinate basis on the tangent spaces of points in the image of ϕ . Then we can represent g by the symmetric matrix $[g_{ij}]_{i,j}$ where

$$g_{ij}(p) = g_p \left(\frac{\partial}{\partial x_i}(p), \frac{\partial}{\partial x_j}(p) \right).$$

In the above I've written g_{ij} as a function on $\phi(U) \subset M$, but it is also common to think of g_{ij} as a function on $U \subset \mathbb{R}^n$: $g_{ij}(x_1, \dots, x_n) = g_{ij}(\phi(x_1, \dots, x_n))$. Of course, the fact that g is a smooth tensor field implies that, in this interpretation, g_{ij} is a smooth function on U .¹

Before giving examples, let's define the appropriate notion of isomorphism of Riemannian metrics.

¹For those who have taken an undergraduate differential geometry course like MATH 474, the matrix $[g_{ij}]_{i,j}$ is the same thing as the matrix usually called the *first fundamental form* in that course.

Definition 4.1.2. Let (M, g_M) and (N, g_N) be Riemannian manifolds. If $f: M \rightarrow N$ is a diffeomorphism so that

$$(g_M)_p(u, v) = (g_N)_{f(p)}(df_p u, df_p v)$$

for all $p \in M$ and all $u, v \in T_p M$, then f is an *isometry*.

A smooth map $f: M \rightarrow N$ is a *local isometry* at $p \in M$ if there exists a neighborhood U of p so that $f: U \rightarrow f(U)$ is an isometry. M and N are *locally isometric* if for every $p \in M$ there exists a neighborhood U of p and a local isometry $f: U \rightarrow f(U)$.

Example 4.1.3. The simplest possible example is $M = \mathbb{R}^n$, where we identify each tangent space $T_p M$ with \mathbb{R}^n so that $\frac{\partial}{\partial x_i}$ gets identified with $e_i = (0, \dots, 0, 1, 0, \dots, 0)$, the i th standard basis vector. Then $g\left(\frac{\partial}{\partial x_i}, \frac{\partial}{\partial x_j}\right) = \delta_{ij}$ (i.e., just the standard dot product on each tangent space) is a Riemannian metric on \mathbb{R}^n , called the *Euclidean metric*.

Example 4.1.4. Suppose $f: M^m \rightarrow N^n$ is an immersion (so $m \leq n$). If N is a Riemannian manifold with Riemannian metric g_N , then f induces a Riemannian metric on M by

$$(g_M)_p(u, v) := (g_N)_{f(p)}(df_p u, df_p v)$$

for all $p \in M$ and all $u, v \in T_p M$. Since f is an immersion, df_p is injective for all $p \in M$, which ensures that g is positive definite. The metric on M is sometimes called the metric *induced* by f and it turns f into an isometry (and therefore an *isometric immersion*).

Example 4.1.5. Let $h: M^m \rightarrow N^n$ be smooth and suppose $q \in N$ is a regular value of h (which necessarily implies that $m \geq n$). We know from [Theorem 1.3.9](#) that $h^{-1}(q) \subset M$ is a smooth submanifold. Hence, if M is Riemannian, then by the previous example we get a metric on $h^{-1}(q)$ induced by the inclusion map $i: h^{-1}(q) \hookrightarrow M$.

For example, if $h: \mathbb{R}^n \rightarrow \mathbb{R}$ is given by $h(x_1, \dots, x_n) := x_1^2 + \dots + x_n^2$ then, as we saw in [Example 1.3.10](#), 1 is a regular value and of course $h^{-1}(1)$ is the unit sphere in \mathbb{R}^n . The metric on the sphere induced by the Euclidean metric is the *standard round metric* on S^{n-1} .

Example 4.1.6. Suppose G is a Lie group with Lie algebra \mathfrak{g} . Choose any inner product $\langle \cdot, \cdot \rangle_e$ on $T_e G \cong \mathfrak{g}$. Then we can define a (left-invariant) metric on G by, for each $h \in G$ and $u, v \in T_h G$,

$$g_h(u, v) := \langle (dL_{h^{-1}})_h u, (dL_{h^{-1}})_h v \rangle_e.$$

Of course, we could just as well have produced a right-invariant metric instead by pushing forward by $dR_{h^{-1}}$.

Example 4.1.7. Recall the affine group $\text{Aff}(\mathbb{R})$ from [Example 3.2.3](#), where $\text{Aff}(\mathbb{R})$ consisted of invertible affine transformations $\phi_{a,b}(t) = at + b$, where $a \neq 0$.

For this example, I'm going to change notation and think of affine transformations $\phi_{x,y}(t) := yt + x$, which is invertible if $y \neq 0$. Let $\text{Aff}^+(\mathbb{R})$ be the subgroup of orientation-preserving affine transformations, meaning that $y > 0$. Then we can identify $\phi_{x,y}$ with the point $(x, y) \in H$, where $H = \{(x, y) \in \mathbb{R}^2 : y > 0\}$ is the upper half-plane. Under this identification, the identity element $e \in \text{Aff}^+(\mathbb{R})$ corresponds to the point $(0, 1) \in H$. Of course, $T_e \text{Aff}^+(\mathbb{R}) \cong T_{(0,1)} \mathbb{R}^2 \cong \mathbb{R}^2$, so we can define an inner product on $T_e \text{Aff}^+(\mathbb{R})$ by simply taking the standard dot product on \mathbb{R}^2 ; i.e.

$$\langle \frac{\partial}{\partial x}, \frac{\partial}{\partial x} \rangle_e = 1, \quad \langle \frac{\partial}{\partial x}, \frac{\partial}{\partial y} \rangle_e = 0, \quad \langle \frac{\partial}{\partial y}, \frac{\partial}{\partial y} \rangle_e = 1,$$

or, in g_{ij} notation, $g_{11}(e) = g_{22}(e) = 1$ and $g_{12}(e) = g_{21}(e) = 0$.

So what is the corresponding left-invariant Riemannian metric on $\text{Aff}^+(\mathbb{R})$? Translating the results of Exercise 3.3.10 to the present notation tells us that

$$(dL_{\phi_{x,y}})_e \frac{\partial}{\partial x} = y \frac{\partial}{\partial x} \quad \text{and} \quad (dL_{\phi_{x,y}})_e \frac{\partial}{\partial y} = y \frac{\partial}{\partial y},$$

or equivalently

$$(dL_{\phi_{x,y}^{-1}})_{\phi_{x,y}} \frac{\partial}{\partial x} = \frac{1}{y} \frac{\partial}{\partial x} \quad \text{and} \quad (dL_{\phi_{x,y}^{-1}})_{\phi_{x,y}} \frac{\partial}{\partial y} = \frac{1}{y} \frac{\partial}{\partial y}.$$

Therefore, by the previous example, the left-invariant metric on $\text{Aff}^+(\mathbb{R})$ is given by

$$\begin{aligned} g_{11}(x, y) &= \left\langle \frac{1}{y} \frac{\partial}{\partial x}, \frac{1}{y} \frac{\partial}{\partial x} \right\rangle_e = \frac{1}{y^2} \\ g_{12}(x, y) &= \left\langle \frac{1}{y} \frac{\partial}{\partial x}, \frac{1}{y} \frac{\partial}{\partial y} \right\rangle_e = 0 \\ g_{22}(x, y) &= \left\langle \frac{1}{y} \frac{\partial}{\partial y}, \frac{1}{y} \frac{\partial}{\partial y} \right\rangle_e = \frac{1}{y^2}. \end{aligned}$$

This is the same as the hyperbolic metric on H discussed in Example 2.3.22!

Exercise 4.1.8. Write $(x, y) = x + iy = z$ for each $(x, y) \in H$. Show that, if $a, b, c, d \in \mathbb{R}$ with $ad - bc = 1$, then

$$z \mapsto \frac{az + b}{cz + d}$$

is an isometry with respect to the metric just defined on H .

I promised above that Riemannian metrics give a notion of length on manifolds. Of course, given a Riemannian metric, I get a notion of lengths of tangent vectors by just taking the norm associated to the inner product: for $v \in T_p M$, its norm/length is just $\sqrt{g_p(v, v)}$.

But a Riemannian metric also gives a notion of lengths of curves by integrating the norm of the tangent vector to the curve:

Definition 4.1.9. Let (M, g) be a Riemannian manifold and let $\alpha: [a, b] \rightarrow M$ be a smooth curve in M . Define the *length* of α in M to be

$$\int_a^b \sqrt{g_{\alpha(t)}(\alpha'(t), \alpha'(t))} dt.$$

Given a notion of length, we also get a notion of volume. First, if we choose some orthonormal basis e_1, \dots, e_n for $T_p M$ and consider the coordinate vectors $X_i(p) := \frac{\partial}{\partial x_i}(p)$, then there exist a_{ij} so that

$$X_i(p) = \sum_{i,j} a_{ij} e_j.$$

The volume $\text{vol}(X_1(p), \dots, X_n(p))$ of the parallelepiped spanned by the X_i in $T_p M$ is equal to $\text{vol}(e_1, \dots, e_n) = 1$ times the determinant of the change-of-basis matrix $A := [a_{ij}]_{i,j}$:

$$\text{vol}(X_1(p), \dots, X_n(p)) = (\det A)1 = \det A. \tag{4.1}$$

To relate this to the metric, observe that

$$g_{ij}(p) = g_p(X_i(p), X_j(p)) = g_p\left(\sum_{k=1}^n a_{ik} e_k, \sum_{\ell=1}^n a_{j\ell} e_\ell\right) = \sum_{k,\ell} a_{ik} a_{j\ell} g_p(e_k, e_\ell) = \sum_{k=1}^n a_{ik} a_{jk},$$

which is just the (i, j) entry of AA^T . Since

$$\det A = \sqrt{\det AA^T} = \sqrt{\det [g_{ij}]_{i,j}}, \quad (4.2)$$

we can combine (4.1) and (4.2) to see that

$$\text{vol}(X_1(p), \dots, X_n(p)) = \sqrt{\det g_{ij}}(p),$$

where $\sqrt{\det g_{ij}}$ is the standard shorthand for $\sqrt{\det [g_{ij}]_{i,j}}$.

Notice that, if p lies in another local coordinate chart (V, ψ) with coordinate basis $Y_i(p) = \frac{\partial}{\partial y_i}(p)$, then in those coordinates the Riemannian metric looks like $[h_{ij}]_{i,j}$ with $h_{ij}(p) := g_p(Y_i(p), Y_j(p))$ and

$$\sqrt{\det g_{ij}}(p) = \text{vol}(X_1(p), \dots, X_n(p)) = (\det J) \text{vol}(Y_1(p), \dots, Y_n(p)) = (\det J) \sqrt{\det h_{ij}}(p), \quad (4.3)$$

where $J = \begin{bmatrix} \frac{\partial x_i}{\partial y_j} \end{bmatrix}_{i,j}$ is the Jacobian of the change of coordinates map $\psi^{-1} \circ \phi: U \rightarrow V$.

Definition 4.1.10. Let (U, ϕ) be a local coordinate chart on a Riemannian manifold (M, g) and assume that $R \subset \phi(U)$ is an open subset with compact closure. Then the *volume* of R is

$$\text{vol}_g(R) := \int_{\phi^{-1}(R)} \sqrt{\det g_{ij}} dx_1 \dots dx_n.$$

This is well-defined: if $R \subset \psi(V)$ for some other local coordinate chart (V, ψ) , then

$$\int_{\psi^{-1}(R)} \sqrt{\det h_{ij}} dy_1 \dots dy_n = \int_{\phi^{-1}(R)} \sqrt{\det g_{ij}} dx_1 \dots dx_n = \text{vol}_g(R)$$

by (4.3) and the change of variables formula.

More generally, we can define the volume of a region not contained in a single coordinate chart using a partition of unity, as in [Definition 2.9.6](#). Indeed, the point is really that a Riemannian metric determines a volume form:

Definition 4.1.11. Let (M, g) be an orientable Riemannian manifold and define the *Riemannian volume form* $d\text{Vol}_g \in \Omega^n(M)$, which in any oriented local coordinate chart has the form

$$d\text{Vol}_g = \sqrt{\det g_{ij}} dx_1 \wedge \dots \wedge dx_n.$$

If $\text{vol}_g(M) = \int_M d\text{Vol}_g$ is finite (e.g., if M is compact), then the Riemannian volume form determines a (signed) measure, called the *Riemannian volume measure*, on the Borel sets in M :

$$\text{vol}_g(B) := \int_B d\text{Vol}_g.$$

(If you don't want a signed measure, just take the absolute value of the right hand side.)

So, for example, if (M, g) is a compact Riemannian manifold, then we automatically get a probability measure on M by normalizing the above measure:

$$P_g(B) := \left| \frac{\text{vol}_g(B)}{\text{vol}_g(M)} \right|.$$

4.2 Affine Connections

We would like to define geodesics on Riemannian manifolds. These are the analogs of straight lines, in the sense both that they are defined as “straightest” paths on Riemannian manifolds and that shortest paths between points are always geodesics. To do so, we will need to define an intrinsic notion of the derivative of a vector field along a curve; then the geodesics will be the curves whose tangent vectors differentiate to zero (just like the tangent vector to a straight line parametrized at constant speed is constant, and hence its derivative is zero; or, if you like, the acceleration along a straight line is zero).

This notion of differentiation will be called the *covariant derivative*. If our manifold is a smooth surface in \mathbb{R}^3 (with the Riemannian metric induced by the Euclidean structure on \mathbb{R}^3), then there's a simple geometric description of the covariant derivative, which I will now try to explain.

Let S be a smooth surface in \mathbb{R}^3 and let $\alpha: I \rightarrow S$ be a smooth curve in S , where I is some interval. Suppose V is a vector field along α . Concretely, we can write $V: I \rightarrow \mathbb{R}^3$ so that $V(t) \in T_{\alpha(t)}S$: that is, at each t , the vector $V(t)$ is tangent to the curve α .

In general, there is no reason that $V'(t)$ should be tangent to S .

Example 4.2.1. Consider the case where S is the unit sphere, $\alpha(t) = \left(\frac{1}{\sqrt{2}} \cos t, \frac{1}{\sqrt{2}} \sin t, \frac{1}{\sqrt{2}} \right)$ is the circle of latitude at 45° North, and

$$V(t) := \left(\sin^2 t - \frac{1}{\sqrt{2}} \cos^2 t, -\frac{2 + \sqrt{2}}{4} \sin 2t, \frac{1}{\sqrt{2}} \cos t \right).$$

See [Figure 4.1](#).

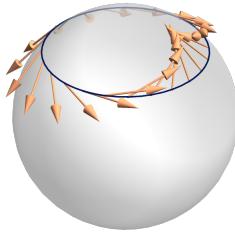


Figure 4.1: The unit sphere with curve $\alpha(t)$ colored in dark blue and the vector field $V(t)$ in orange.

Then

$$V'(t) = \left((2 + \sqrt{2}) \cos t \sin t, -\frac{2 + \sqrt{2}}{2} \cos 2t, -\frac{\sin t}{\sqrt{2}} \right).$$

Hence $V'(\pi/2) = \left(0, 1 + \frac{1}{\sqrt{2}}, -\frac{1}{\sqrt{2}}\right)$ is not in the tangent space to $\alpha(\pi/2) = \left(0, \frac{1}{\sqrt{2}}, \frac{1}{\sqrt{2}}\right)$ since

$$V'(\pi/2) \cdot \alpha(\pi/2) = \frac{1}{\sqrt{2}} \neq 0.$$

The fact that $V'(t)$ is not generally tangent to S means that $V'(t)$ is not a notion of derivative that can be defined intrinsically to the surface. However, in this case, the simplest possible fix turns out to give something that *is* intrinsically defined. Specifically, we know that $\alpha(t) \in S$, and so we have the tangent space $T_{\alpha(t)}S$ to S at $\alpha(t)$. Of course, $\alpha(t) \in \mathbb{R}^3$ as well, and \mathbb{R}^3 has its own tangent space at $\alpha(t)$, namely $T_{\alpha(t)}\mathbb{R}^3$. But then $V'(t) \in T_{\alpha(t)}\mathbb{R}^3$ and $T_{\alpha(t)}S \subset T_{\alpha(t)}\mathbb{R}^3$, so we can orthogonally project $V'(t)$ to $T_{\alpha(t)}S$, as in Figure 4.2.

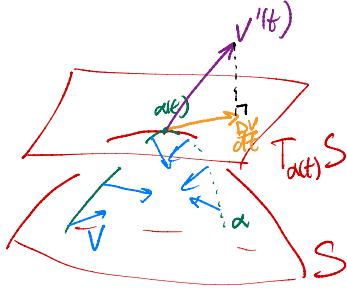


Figure 4.2: The projection of $V'(t)$ onto $T_{\alpha(t)}S$ yields the covariant derivative $\frac{DV}{dt}$.

The projection of $V'(t)$ onto $T_{\alpha(t)}S$ is called the *covariant derivative* of V along α and denoted $\frac{DV}{dt}$. While this still seems extrinsic to the surface, it turns out you can define $\frac{DV}{dt}$ purely in terms of α , V , S , and the Riemannian metric g on S . While in this example the Riemannian metric is induced by the Euclidean metric on \mathbb{R}^3 , you can use the same formula to define $\frac{DV}{dt}$ for any Riemannian metric g on a surface, regardless of whether or not g is induced from a Euclidean metric.

In turn, we say that a vector field V along α is *parallel* if $\frac{DV}{dt} \equiv 0$; then a geodesic is just a curve α so that $\alpha'(t)$ is parallel along α .

If you want to see more details on the above, see any standard textbook on curves and surfaces, for example do Carmo's book [6] or O'Neill's [23]. Here, I'm going to develop the general theory on arbitrary Riemannian manifolds, which uses a gadget called a *connection*. These can be defined on any manifold (not just Riemannian manifolds), but, as we'll see, a Riemannian metric produces a special connection.

Definition 4.2.2. An *affine connection* ∇ on a manifold M is a mapping

$$\nabla: \mathfrak{X}(M) \times \mathfrak{X}(M) \rightarrow \mathfrak{X}(M)$$

sending $(V, W) \mapsto \nabla_V W$ satisfying the following:

- (i) Linearity in the first factor: $\nabla_{f_1 V_1 + f_2 V_2} W = f_1 \nabla_{V_1} W + f_2 \nabla_{V_2} W$
- (ii) Additivity in the second factor: $\nabla_V (W_1 + W_2) = \nabla_V W_1 + \nabla_V W_2$
- (iii) The Leibniz rule in the second factor: $\nabla_V (f W) = V(f)W + f \nabla_V W$.

Example 4.2.3. In \mathbb{R}^n , suppose $W = \sum_k w_k \frac{\partial}{\partial x_k}$ and define

$$\nabla_V W = \sum_k V(w_k) \frac{\partial}{\partial x_k}.$$

Lemma 4.2.4. *Affine connections are local in the sense that the value of $\nabla_V W$ at p depends only on $V(p)$ and the values of W in a neighborhood of p .*

Proof. Let g be a smooth function with support in some small neighborhood of p , with $g \equiv 1$ on some even smaller neighborhood. Then at p we have $V(g) = 0$, and hence at p

$$\nabla_V(gW) = V(g)W + g\nabla_V W = \nabla_V W$$

by (iii). Since the support of the cutoff function g can be arbitrarily small, we see that $\nabla_V W$ depends only on the values of W in a small neighborhood of p .

From (i), we know $\nabla_{fV} W = f\nabla_V W$, so it follows immediately that $\nabla_V W$ at p also depends only on the values of V in a neighborhood of p .

In fact, we can show more. Let (U, ϕ) be a coordinate chart in a neighborhood of p and let $X_1 = \frac{\partial}{\partial x_1}, \dots, X_n = \frac{\partial}{\partial x_n}$ be the coordinate vector fields. If $V = \sum_i v_i X_i$, then

$$\nabla_V W = v_1 \nabla_{X_1} W + \dots + v_n \nabla_{X_n} W$$

by (i). If we evaluate the right hand side at p , the only dependence on V come from the $v_i(p)$, so it only depends on $V(p)$ (and not on the behavior of V in a neighborhood of p). \square

If ∇ is an affine connection on M and $X_1 = \frac{\partial}{\partial x_1}, \dots, X_n = \frac{\partial}{\partial x_n}$ in some coordinate neighborhood on M , define the smooth functions Γ_{ij}^k on that coordinate neighborhood by

$$\nabla_{X_i} X_j = \sum_k \Gamma_{ij}^k X_k.$$

The Γ_{ij}^k are called the *Christoffel symbols* of the connection ∇ , and we can write a formula for $\nabla_V W$ in terms of them: if $V = \sum_i v_i X_i$ and $W = \sum_j w_j X_j$ are vector fields defined on a coordinate neighborhood, then

$$\begin{aligned} \nabla_V W &= \sum_i v_i \nabla_{X_i} \left(\sum_j w_j X_j \right) \\ &= \sum_{i,j} (v_i X_i(w_j) X_j + v_i w_j \nabla_{X_i} X_j) \\ &= \sum_{i,j} \left(v_i X_i(w_j) X_j + v_i w_j \sum_k \Gamma_{ij}^k X_k \right) \\ &= \sum_{i,j,k} \left(v_i X_i(w_k) + v_i w_j \Gamma_{ij}^k \right) X_k \\ &= \sum_{i,j,k} \left(v_i \frac{\partial w_k}{\partial x_i} + v_i w_j \Gamma_{ij}^k \right) X_k. \end{aligned}$$

Exercise 4.2.5. Show that, within a coordinate neighborhood, if we pick n^3 smooth functions Γ_{ij}^k arbitrarily and define

$$\nabla_V W := \sum_{i,j,k} \left(v_i \frac{\partial w_k}{\partial x_i} + v_i w_j \Gamma_{ij}^k \right) X_k,$$

then ∇ will satisfy the axioms of an affine connection in that coordinate neighborhood.

Remark 4.2.6. Despite the suggestive notation, the Γ_{ij}^k are not the components of a tensor field on M : they do not transform like tensor components under general coordinate transformations.

We can refine [Lemma 4.2.4](#) a bit: we don't actually need to know W on an entire neighborhood of p , but only along a curve within that neighborhood:

Lemma 4.2.7. *The value of $\nabla_V W$ at p depends only on the value of V at p and the value of W along any (short) curve through p which is tangent there to $V(p)$.*

Proof. We've just seen that, in local coordinates,

$$\nabla_V W = \sum_{i,j,k} \left(v_i \frac{\partial w_k}{\partial x_i} + v_i w_j \Gamma_{ij}^k \right) X_k = \sum_{i,j,k} \left(V(w_k) + v_i w_j \Gamma_{ij}^k \right) X_k \quad (4.4)$$

since $V = \sum_i v_i \frac{\partial}{\partial x_i}$. The only term which is not entirely determined by $V(p)$ and $W(p)$ is $V(w_k) = (\alpha \circ w_k)'(0)$ for some curve α with $\alpha(0) = p$ and $\alpha'(0) = V(p)$. Of course, $(\alpha \circ w_k)(t)$ only depends on the values of W along α . \square

An affine connection then gives us a way of defining covariant derivatives:

Proposition 4.2.8. *Let ∇ be an affine connection on a smooth manifold M . Then we can uniquely associate to any vector field W along a curve $\alpha: I \rightarrow M$ another vector field $\frac{DW}{dt}$ along α such that*

$$(i) \quad \frac{D(W_1+W_2)}{dt} = \frac{DW_1}{dt} + \frac{DW_2}{dt}$$

$$(ii) \quad \frac{DgW}{dt} = \frac{dg}{dt} W + g \frac{DW}{dt}$$

(iii) *If W is defined on all of M , then*

$$\frac{DW}{dt} = \nabla_{\frac{d\alpha}{dt}} W.$$

Definition 4.2.9. The vector field $\frac{DW}{dt}$ guaranteed by [Proposition 4.2.8](#) is called the *covariant derivative* of W along α .

Proof of Proposition 4.2.8. If we work in local coordinates, we have $\alpha(t) = (\alpha_1(t), \dots, \alpha_n(t))$, so

$$\frac{d\alpha}{dt} = \sum_i \frac{d\alpha_i}{dt} X_i,$$

so the formula (4.4) for $\nabla_V W$ shows that we must define

$$\frac{DW}{dt} := \sum_{i,j,k} \left(\frac{dw_k}{dt} + \frac{d\alpha_i}{dt} w_j \Gamma_{ij}^k \right) X_k \quad (4.5)$$

in order to satisfy conditions (i), (ii), and (iii). Hence, the covariant derivative, if it exists, must be unique. But to get existence we just define the covariant derivative in coordinates as in (4.5). \square

The thing to remember about affine connections is that they give a way of defining parallel transport; that is, some way of moving a tangent vector “parallel to itself” along a curve. In general, there’s no canonical way to do this, so you have to define what you mean by “parallel,” and that’s really what a connection does.

Example 4.2.10. Consider the unit sphere $S^2 \subset \mathbb{R}^3$ and take the tangent vector $\frac{\partial}{\partial y}$ at the north pole. Say we want to transport this vector parallel to itself around the spherical triangle that traverses a great circle south from $(0, 0, 1)$ to $(1, 0, 0)$, then the equator east from $(1, 0, 0)$ to $(0, 1, 0)$, and finally another great circle north from $(0, 1, 0)$ to $(0, 0, 1)$. Of course, we could move $\frac{\partial}{\partial y}$ parallel to itself in \mathbb{R}^3 , which just gives $\frac{\partial}{\partial y}$ at every point along the curve; see [Figure 4.3](#). But $\frac{\partial}{\partial y}$ is not in the tangent space at $(0, 1, 0) \in S^2$: it’s perpendicular to the sphere at that point!

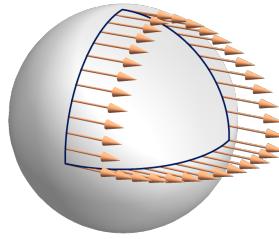


Figure 4.3: Parallel translating a vector around a curve in the ambient space.

To get something which stays tangent to the sphere, we simply take the vector field which is always pointing east. Along the first segment, traveling south from the north pole to $(1, 0, 0)$, this is the same thing we had before. You can imagine a traveler walking south along this path, and then the vector field is just what they get when they stick out their left hand.

At $(1, 0, 0)$, the traveler turns left, and now east is straight ahead as they start walking toward $(0, 1, 0)$, so now the vector field is what they get by pointing straight ahead. From the perspective of the traveler, this is the same direction as along the first leg of their journey (namely, east!).

When they arrive at $(0, 1, 0)$, the traveler now turns left again to face north. Now east is to their right, and they can trace out the vector field by just sticking out their right hand as they walk north. Notice that, once the traveler gets back to the north pole, the vector we get by parallel translating the initial vector around the curve is perpendicular to the initial vector! So even though all along the way the vector field was constant (from the perspective of the traveler or, more mathematically, with respect to the intrinsic geometry of the sphere), it realizes some nontrivial transformation of the initial vector.² See [Figure 4.4](#).

The idea of parallel translation is the same in general: the idea is that, as you traverse a curve, you’re not changing the direction of the vector.

Definition 4.2.11. Let M be a smooth manifold with affine connection ∇ . A vector field V along a curve $\alpha: I \rightarrow M$ is *parallel* (along α) if $\frac{dV}{dt} = 0$ for all $t \in I$.

Example 4.2.12. With respect to the standard connection ∇ on \mathbb{R}^n from [Example 4.2.3](#), a vector field $V = \sum_i v_i \frac{\partial}{\partial x_i}$ is parallel (along any curve) if and only if the v_i are constants. The V shown in [Figure 4.3](#) is an example if you just think of the curve as a curve in \mathbb{R}^3 and ignore the sphere.

²This phenomenon is called *holonomy*, and arises whenever there is curvature.

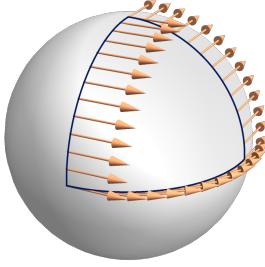


Figure 4.4: Parallel translating a vector around a curve intrinsically.

It turns out that, given any initial vector at a point and any curve through the point, we can always find a parallel field along the curve which agrees with the given vector at the point.

Proposition 4.2.13. *Let M be a smooth manifold with an affine connection ∇ , let $\alpha: I \rightarrow M$ be a smooth curve in M , and let $V_0 \in T_{\alpha(t_0)}M$ be some tangent vector to M at $\alpha(t_0)$. Then there is a unique parallel vector field $V: I \rightarrow TM$ along α so that $V(t) \in T_{\alpha(t)}M$ for all $t \in I$ and $V(t_0) = V_0$.*

Proof. In local coordinates, the parallel condition $\frac{DV}{dt} = 0$ is a system of n first order linear ODEs for the coefficients of V . Then the standard existence and uniqueness theorems apply and we get a unique solution with given initial condition. \square

Definition 4.2.14. The vector field $V(t)$ guaranteed by Proposition 4.2.13 is called the *parallel transport* of $V_0 = V(t_0)$ along the curve $\alpha(t)$.

From here, we could define geodesics, curvatures, etc. for arbitrary connections. Rather than develop this theory in full generality, we will specialize to the case of the *Riemannian connection* or *Levi-Civita connection*, which is uniquely determined by a Riemannian metric. So we turn to the task of defining this special connection.

4.3 The Levi-Civita Connection

We need to define two notions, compatibility and symmetry, before we can give the main theorem of this section, which is the existence of the Levi-Civita connection.

Definition 4.3.1. Let (M, g) be a Riemannian manifold with affine connection ∇ . The connection is called *compatible* with g if, for any smooth curve α and any pair of parallel vector fields V and W along α , the function $t \mapsto g_{\alpha(t)}(V(t), W(t))$ is constant along α .

Example 4.3.2. The Euclidean connection on \mathbb{R}^n from Example 4.2.3 is compatible with the Euclidean metric since parallel vector fields are constant.

Exercise 4.3.3. Show that the restriction of the Euclidean connection to a submanifold $M^m \subset \mathbb{R}^n$ is compatible with the metric on M induced by the Euclidean metric on \mathbb{R}^n .

Exercise 4.3.4. Show that an affine connection ∇ is compatible with a Riemannian metric g on M if and only if, for any vector fields V and W along a smooth curve $\alpha: I \rightarrow M$, we have

$$\frac{d}{dt} \Big|_{t=t_0} g_{\alpha(t)}(V(t), W(t)) = g_{\alpha(t_0)} \left(\frac{DV}{dt}, W \right) + g_{\alpha(t_0)} \left(V, \frac{DW}{dt} \right).$$

In other words, for compatible connections we can use the usual product rule to differentiate the inner product.

Corollary 4.3.5. Let (M, g) be a Riemannian manifold. An affine connection ∇ on M is compatible with g if and only if

$$U(g(V, W)) = g(\nabla_U V, W) + g(V, \nabla_U W) \quad (4.6)$$

at all points of M and for all $U, V, W \in \mathfrak{X}(M)$.

Proof. Suppose ∇ is compatible with g . Let $p \in M$ and let $\alpha: (-\epsilon, \epsilon) \rightarrow M$ be a smooth curve with $\alpha(0) = p$ and $\alpha'(0) = U(p)$. Then

$$\begin{aligned} U(g(V, W))(p) &= \frac{d}{dt} \Big|_{t=t_0} g_{\alpha(t)}(V, W) \\ &= g_p \left(\frac{DV}{dt}, W \right) + g_p \left(V, \frac{DW}{dt} \right) \\ &= g_p(\nabla_{U(p)} V, W(p)) + g_p(V(p), \nabla_{U(p)} W), \end{aligned}$$

where the second line follows from Exercise 4.3.4 and the third from Proposition 4.2.8(iii). Since p was chosen arbitrarily, (4.6) follows.

On the other hand, if ∇ satisfies (4.6) and V, W are parallel along $\alpha: I \rightarrow M$, then Exercise 4.3.4 implies that

$$\frac{d}{dt} g_{\alpha(t)}(V(t), W(t)) = g_{\alpha(t)} \left(\frac{DV}{dt}, W \right) + g_{\alpha(t)} \left(V, \frac{DW}{dt} \right) = 0$$

since $\frac{DV}{dt} \equiv 0$ and $\frac{DW}{dt} \equiv 0$. □

Definition 4.3.6. A connection ∇ on M is called *symmetric* or *torsion-free* if

$$\nabla_V W - \nabla_W V = [V, W]$$

for all $V, W \in \mathfrak{X}(M)$.

Example 4.3.7. Consider the Euclidean connection on \mathbb{R}^n again (as defined in Example 4.2.3). We use the standard coordinates on \mathbb{R}^n , and assume $V = \sum_i v_i \frac{\partial}{\partial x_i}$ and $W = \sum_j w_j \frac{\partial}{\partial x_j}$. Then

$$\begin{aligned} \nabla_V W &= \sum_{i,j} v_i \frac{\partial w_j}{\partial x_i} \frac{\partial}{\partial x_j} \\ \nabla_W V &= \sum_{i,j} w_j \frac{\partial v_i}{\partial x_j} \frac{\partial}{\partial x_i}, \end{aligned}$$

so we can match indices and conclude that

$$\nabla_V W - \nabla_W V = \sum_{i,j} \left(v_i \frac{\partial w_j}{\partial x_i} - w_i \frac{\partial v_j}{\partial x_i} \right) \frac{\partial}{\partial x_j}.$$

On the other hand, we saw way back in (1.3) that this is the local coordinate expression for $[V, W]$, so we see that the Euclidean connection is symmetric.

Example 4.3.8. If we work in a local coordinate chart on a manifold M and write $X_i = \frac{\partial}{\partial x_i}$, then ∇ being symmetric implies

$$\nabla_{X_i} X_j - \nabla_{X_j} X_i = [X_i, X_j] = 0;$$

that is,

$$\nabla_{X_i} X_j = \nabla_{X_j} X_i.$$

In fact, it's not hard to show that ∇ is symmetric if and only if the above equation holds; this explains the terminology *symmetric*.

Since the Christoffel symbols are defined by

$$\nabla_{X_i} X_j = \sum_k \Gamma_{ij}^k X_k,$$

the connection ∇ is symmetric if and only if

$$\Gamma_{ij}^k = \Gamma_{ji}^k$$

for all i, j, k . In turn, we know from [Exercise 4.2.5](#) that any choice of the Christoffel symbols Γ_{ij}^k determines a connection, so we can construct all symmetric connections by choosing $n \binom{n+1}{2} = \frac{n^2(n+1)}{2}$ functions Γ_{ij}^k so that $\Gamma_{ij}^k = \Gamma_{ji}^k$ for all i, j, k .

We are now ready to state and prove the main result:

Theorem 4.3.9 (Fundamental Theorem of Riemannian Geometry; Levi-Civita [[20](#)]). *Let (M, g) be a Riemannian manifold. There exists a unique affine connection ∇ so that*

(i) ∇ is symmetric and

(ii) ∇ is compatible with g .

This connection is usually called the *Levi-Civita connection* or the *Riemannian connection*.

Proof. First we prove uniqueness. So suppose there were some such connection ∇ . Then, for any $U, V, W \in \mathfrak{X}(M)$, [Corollary 4.3.5](#) implies

$$U(g(V, W)) = g(\nabla_U V, W) + g(V, \nabla_U W) \tag{4.7}$$

$$V(g(W, U)) = g(\nabla_V W, U) + g(W, \nabla_V U) \tag{4.8}$$

$$W(g(U, V)) = g(\nabla_W U, V) + g(U, \nabla_W V). \tag{4.9}$$

Since the connection is symmetric, we know that

$$\nabla_U V - \nabla_V U = [U, V]$$

and similarly for other terms, so subtracting (4.9) from the sum of (4.7) and (4.8) yields

$$\begin{aligned} U(g(V, W)) + V(g(W, U)) - W(g(U, V)) &= g([U, W], V) + g([V, W], U) + g(\nabla_U V + \nabla_V U, W) \\ &= g([U, W], V) + g([V, W], U) + g([U, V], W) + 2g(\nabla_V U, W). \end{aligned}$$

In other words,

$$g(\nabla_V U, W) = \frac{1}{2} [U(g(V, W)) + V(g(W, U)) - W(g(U, V)) - g([U, W], V) - g([V, W], U) - g([U, V], W)], \quad (4.10)$$

which shows that ∇ is uniquely determined by g . Hence, if it exists, ∇ must be unique.³

To prove existence, we can simply define ∇ by Equation (4.10); it is straightforward to check that ∇ defined in this way is a symmetric, compatible connection. \square

Example 4.3.10. Let's work out the Levi-Civita connection for the Riemannian metric on $\text{Aff}^+(\mathbb{R})$ that we defined in Example 4.1.7. Recall that we can identify $\text{Aff}^+(\mathbb{R})$ with the upper half-plane $H = \{(x, y) \in \mathbb{R}^2 : y > 0\}$ and that the metric g can be represented by the matrix

$$g_{(x,y)} \leftrightarrow \begin{bmatrix} g_{11}(x, y) & g_{12}(x, y) \\ g_{12}(x, y) & g_{22}(x, y) \end{bmatrix} = \begin{bmatrix} \frac{1}{y^2} & 0 \\ 0 & \frac{1}{y^2} \end{bmatrix}.$$

Let $X = \frac{\partial}{\partial x}$ and $Y = \frac{\partial}{\partial y}$. Since these are coordinate fields, we know $[X, Y] = 0$, and we always have $[X, X] = 0$ and $[Y, Y] = 0$ for any vector fields. So half of the terms in (4.10) vanish.

The matrix for g is telling us that

$$g_{(x,y)}(X, X) = \frac{1}{y^2}, \quad g_{(x,y)}(X, Y) = 0, \quad g_{(x,y)}(Y, Y) = \frac{1}{y^2},$$

and hence that, using (4.10)

$$\begin{aligned} g_{(x,y)}(\nabla_X X, X) &= \frac{1}{2} [X(g_{(x,y)}(X, X)) + X(g_{(x,y)}(X, X)) - X(g_{(x,y)}(X, X))] \\ &= \frac{1}{2} X(g_{(x,y)}(X, X)) \\ &= \frac{1}{2} \frac{\partial}{\partial x} \left(\frac{1}{y^2} \right) \\ &= 0 \end{aligned}$$

and

$$\begin{aligned} g_{(x,y)}(\nabla_X X, Y) &= \frac{1}{2} [X(g_{(x,y)}(X, Y)) + X(g_{(x,y)}(Y, X)) - Y(g_{(x,y)}(X, X))] \\ &= \frac{1}{2} \left[X(0) + X(0) - Y \left(\frac{1}{y^2} \right) \right] \\ &= -\frac{1}{2} \frac{\partial}{\partial y} \left(\frac{1}{y^2} \right) \\ &= \frac{1}{y^3}. \end{aligned}$$

³An alternative argument works in coordinates and gives explicit equations for the Γ_{ij}^k in terms of the components g_{ij} for the metric. This is directly analogous to the standard formulas for the Christoffel symbols on surfaces.

Therefore, for any a and b ,

$$g_{(x,y)}(\nabla_X X, aX + bY) = \frac{b}{y^3}. \quad (4.11)$$

On the other hand, $\nabla_X X = cX + dY$ for some c and d , and we know that

$$g_{(x,y)}(cX + dY, aX + bY) = \frac{ac}{y^3} + \frac{bd}{y^3}. \quad (4.12)$$

Equating (4.11) and (4.12), we see that $\nabla_X X = \frac{1}{y}Y$.

If we play the same game with $\nabla_Y Y$ and with $\nabla_X Y = \nabla_Y X$ (using the symmetry of ∇ and the fact that $[X, Y] = 0$), we conclude that

$$\nabla_X X = \frac{1}{y}Y, \quad \nabla_X Y = \nabla_Y X = -\frac{1}{y}X, \quad \nabla_Y Y = -\frac{1}{y}Y. \quad (4.13)$$

Now, let's think about parallel transporting a vector along the curve $\alpha(t) = (t, 1)$. Notice that $\alpha'(t) = X$ for all t , so by [Proposition 4.2.8\(iii\)](#), we have that the covariant derivative along α is given by

$$\frac{DW}{dt} = \nabla_{\alpha'(t)} W = \nabla_X W$$

for any tangent vector W . So if we think of a vector field $W(t) = a(t)X + b(t)Y \in T_{\alpha(t)}H = T_{(t,1)}H$, then the parallel transport equation becomes

$$\begin{aligned} 0 = \frac{DW}{dt} &= \nabla_X(a(t)X + b(t)Y) = a'(t)X + a(t)\nabla_X X + b'(t)Y + b(t)\nabla_X Y \\ &= a'(t)X + a(t)Y + b'(t)Y - b(t)X = (a'(t) - b(t))X + (b'(t) + a(t))Y \end{aligned}$$

since $y = 1$ along the whole curve and hence $\frac{1}{y} = 1$. Of course, the coefficients have to be 0, so we get the homogeneous system of ODEs

$$\begin{aligned} a'(t) - b(t) &= 0 \\ b'(t) + a(t) &= 0. \end{aligned}$$

Solutions are of the form

$$\begin{aligned} a(t) &= C_1 \cos t + C_2 \sin t \\ b(t) &= -C_1 \sin t + C_2 \cos t. \end{aligned}$$

[Figure 4.5](#) shows the parallel transports of $X \in T_{(0,1)}H$ and $Y \in T_{(0,1)}H$ along this curve.

Recall that a parallel vector field is always pointing in the same direction from the perspective of someone inside the manifold, so the way to interpret this is that a person traversing the curve $\alpha(t)$ is constantly turning to the left to stay on α .

On the other hand, if $\beta(t) = (0, t)$, then $\beta'(t) = Y$ for all t , and the parallel transport equation is

$$\begin{aligned} 0 = \frac{DW}{dt} &= \nabla_Y(a(t)X + b(t)Y) = a'(t)X + a(t)\nabla_Y X + b'(t)Y + b(t)\nabla_Y Y \\ &= a'(t)X - \frac{a(t)}{t}X + b'(t)Y - \frac{b(t)}{t}Y = \left(a'(t) - \frac{a(t)}{t}\right)X \left(b'(t) - \frac{b(t)}{t}\right)Y \end{aligned}$$

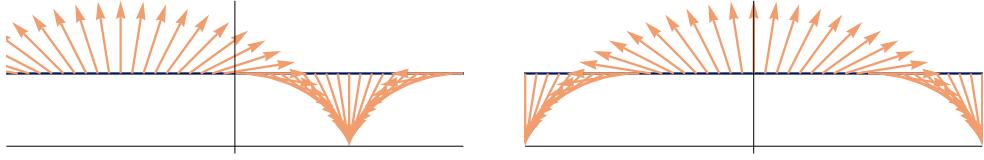


Figure 4.5: The parallel transports of $X \in T_{(0,1)}H$ (left) and $Y \in T_{(0,1)}H$ (right) along the curve $\alpha(t) = (t, 1)$.

since $y = t$ along the curve, and hence $\frac{1}{y} = \frac{1}{t}$. So the system of ODEs is

$$\begin{aligned} a'(t) - \frac{a(t)}{t} &= 0 \\ b'(t) - \frac{b(t)}{t} &= 0. \end{aligned}$$

Solutions are of the form

$$\begin{aligned} a(t) &= C_1 t \\ b(t) &= C_2 t. \end{aligned}$$

Figure 4.6 shows the parallel transports of $X \in T_{(0,1)}H$ and $Y \in T_{(0,1)}H$ along this curve.

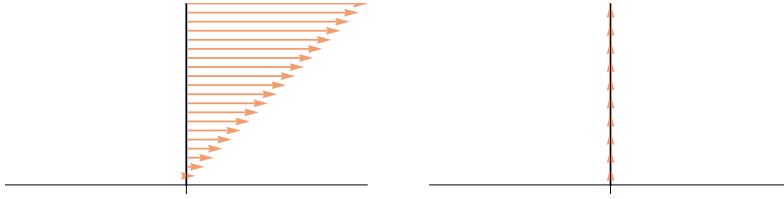


Figure 4.6: The parallel transports of $X \in T_{(0,1)}H$ (left) and $Y \in T_{(0,1)}H$ (right) along the curve $\beta(t) = (0, t)$.

Notice that the parallel transport of a tangent vector to β stays tangent to β ; this is an indication that (up to reparametrization) β is a geodesic.

4.4 Geodesics

Geodesics are generalizations of straight lines to Riemannian manifolds.⁴ Typically we think of straight lines in the plane or in space as *shortest paths*, but this is actually the wrong notion to generalize. Not because shortest paths won't be geodesics, but because not all geodesics are shortest paths.

Example 4.4.1. The geodesics on the round sphere will turn out to be arcs of great circles. Of course, the shortest path between two points will be an arc of a great circle, but we can get another geodesic between

⁴As mentioned briefly at the end of Section 4.2, we can define geodesics for any affine connection, so geodesics are really generalizations of straight lines to manifolds-with-connections, but we're only going to develop the theory here for the Levi-Civita connection.

any two non-antipodal points by going the long way around: instead of traveling from Ecuador to Kenya by flying east along the equator over South America and the Atlantic Ocean, head west over the Pacific Ocean, Indonesia, and the Indian Ocean. The latter route is still a geodesic, but certainly not the shortest path.

Instead, the right notion to generalize is that straight lines are (no surprise) *straightest paths*. We can see this in terms of formulas by taking a constant-speed parametrization of a straight line, something like $\alpha(t) = p + tv$, for $p, v \in \mathbb{R}^n$. Then $\alpha'(t) = v$ is constant, and hence $\alpha''(t) = 0$. The equation $\alpha''(t) = 0$ is going to generalize to the so-called *geodesic equation*.

Definition 4.4.2. Let (M, g) be a Riemannian manifold and let ∇ be the Levi-Civita connection on M . A curve $\gamma: I \rightarrow M$ is a *geodesic* if

$$\frac{D}{dt}\gamma'(t) = 0 \quad (4.14)$$

for all $t \in I$. The equation (4.14) is the *geodesic equation*.

Remark 4.4.3. A geodesic is automatically traversed at constant speed:

$$\frac{d}{dt}g_{\gamma(t)}(\gamma'(t), \gamma'(t)) = 2g_{\gamma(t)}\left(\frac{D}{dt}\gamma'(t), \gamma'(t)\right) = 0$$

by [Exercise 4.3.4](#).

Let's work out what the geodesic equation (4.14) looks like in local coordinates. So work in a local coordinate chart on M . Suppose $\gamma(t) = (x_1(t), \dots, x_n(t))$ is a curve and $W(t) = \sum_j w_j(t) \frac{\partial}{\partial x_j}$ is a vector field along γ . We saw in (4.4) that

$$\frac{DW}{dt} = \sum_{i,j,k} \left(w'_k(t) + \Gamma_{ij}^k x'_i(t) w_j(t) \right) \frac{\partial}{\partial x_k}.$$

So if

$$W(t) = \gamma'(t) = \sum_j x'_j(t) \frac{\partial}{\partial x_j},$$

then we have

$$\frac{D}{dt}\gamma'(t) = \sum_{i,j,k} \left(x'_k(t) + \Gamma_{ij}^k x'_i(t) x'_j(t) \right) \frac{\partial}{\partial x_k}.$$

Then γ is a geodesic if and only if the above is zero, which means all the coefficients must be zero:

$$x''_k(t) + \sum_{i,j} \Gamma_{ij}^k x'_i(t) x'_j(t) = 0 \quad \text{for } k = 1, \dots, n, \quad (4.15)$$

which is a system of n second-order ODEs on a coordinate neighborhood of M .

Example 4.4.4. Let's figure out the geodesic equation on $\text{Aff}^+(\mathbb{R}) \cong H = \{(x, y) \in \mathbb{R}^2 : y > 0\}$. We saw in [Example 4.1.7](#) that the metric is given by $g_{11} = \frac{1}{y^2} = g_{22}$, $g_{12} = g_{21} = 0$, and we worked out the Levi-Civita connection in [Example 4.3.10](#). Specifically, (4.13) implies that

$$\begin{aligned} \Gamma_{11}^1 &= 0 & \Gamma_{12}^1 = \Gamma_{21}^1 &= -\frac{1}{y} & \Gamma_{22}^1 &= 0 \\ \Gamma_{11}^2 &= \frac{1}{y} & \Gamma_{12}^2 = \Gamma_{21}^2 &= 0 & \Gamma_{22}^2 &= -\frac{1}{y}. \end{aligned}$$

Therefore, (4.15) becomes

$$\begin{aligned} x''(t) - \frac{2}{y}x'(t)y'(t) &= 0 \\ y''(t) + \frac{1}{y}((x'(t))^2 - (y'(t))^2) &= 0. \end{aligned} \tag{4.16}$$

On vertical rays, $x(t) = \text{constant}$, so the first equation is satisfied and the second reduces to $y'' = \frac{(y')^2}{y}$, which has solution $y(t) = c_1 e^{c_2 t}$. So the vertical rays parametrized as $\gamma(t) = (x_0, c_1 e^{c_2 t})$ are geodesics. We can eliminate the constants by deciding that we want $\gamma(0) = (a, 1)$, which implies $c_1 = 1$, and that we want $\gamma'(0) = (0, 1)$:

$$(0, 1) = \gamma'(0) = (0, c_2)$$

so $c_2 = 1$. That is, $\gamma(t) = (a, e^t)$ is a geodesic in H for any $a \in \mathbb{R}$.

As for the other geodesics, note that $\frac{dy}{dx} = \frac{y'}{x'}$, which we can differentiate as follows:

$$\frac{d^2y}{dx^2} = \frac{d}{dx} \left(\frac{y'}{x'} \right) = \frac{x'y'' - y'x''}{(x')^2} \frac{1}{x'}$$

since $\frac{df}{dx} = \frac{df}{dt} \frac{dt}{dx} = \frac{\frac{df}{dt}}{\frac{dt}{dx}}$. If we assume our curve $\gamma(t) = (x(t), y(t))$ is a geodesic, we can substitute in $x'' = \frac{2}{y}x'y'$ and $y'' = \frac{1}{y}((y')^2 - (x')^2)$ from (4.16):

$$\begin{aligned} \frac{d^2y}{dx^2} &= \frac{1}{(x')^3} \left(\frac{x'}{y}((y')^2 - (x')^2) - \frac{2}{y}x'(y')^2 \right) \\ &= \frac{1}{(x')^3} \frac{1}{y} \left(-(x')^3 - x'(y')^2 \right) \\ &= -\frac{1}{y} \left(1 + \frac{(y')^2}{(x')^2} \right) \\ &= -\frac{1}{y} \left(1 + \left(\frac{dy}{dx} \right)^2 \right). \end{aligned}$$

(Note that we're implicitly assuming $x' \neq 0$ here, which is why we dealt with the $x(t) = \text{constant}$ case separately.)

In other words,

$$-1 = y \frac{d^2y}{dx^2} + \left(\frac{dy}{dx} \right)^2 = \frac{d}{dx} \left(y \frac{dy}{dx} \right)$$

on a geodesic. Integrating twice gives the solution

$$x^2 + y^2 = ax + b,$$

which is the semi-circle of radius $\sqrt{b + \frac{a^2}{4}}$ centered at the point $(a/2, 0)$.

In other words, we've proved that the geodesics in H are the vertical rays and the semi-circles centered at points on the x -axis, as claimed in Example 2.3.22.

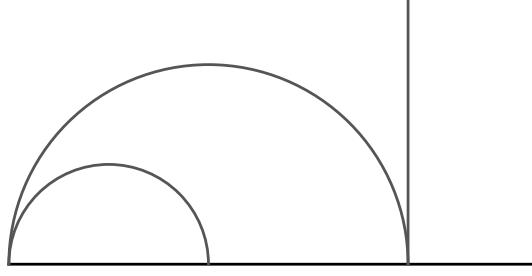


Figure 4.7: Some geodesics in H .

If we want to convert (4.15) to a system of $2n$ first-order ODEs, we need to pass to the tangent bundle TM . Recall that

$$TM := \bigsqcup_{p \in M} T_p M.$$

Suppose $p \in M$ lies in a coordinate chart (U, ϕ) . If $v \in T_p M$, then $v = \sum_i y_i \frac{\partial}{\partial x_i}$ for some y_1, \dots, y_n ; then $(x_1, \dots, x_n, y_1, \dots, y_n)$ give local coordinates on $T(\phi(U)) \cong U \times \mathbb{R}^n$; continuing down this path gives the proof of [Exercise 1.2.6](#).

In any case, the curve $t \mapsto \gamma(t)$ in M determines a curve $t \mapsto (\gamma(t), \gamma'(t))$ in TM and, if γ is a geodesic, then in a local coordinate neighborhood $T(\phi(U))$ this curve satisfies the first-order system

$$\begin{aligned} x'_k(t) &= y_k(t) \\ y'_k(t) &= - \sum_{i,j} \Gamma_{ij}^k y_i(t) y_j(t) \end{aligned} \tag{4.17}$$

for all $k = 1, \dots, n$.

By the usual existence and uniqueness theorems for solutions to ODEs, it will follow that, for any $p \in M$ and any $v \in T_p M$ (or, alternatively, any $(p, v) \in TM$), there will be a geodesic $\gamma: (-\epsilon, \epsilon) \rightarrow M$ with $\gamma(0) = p$ and $\gamma'(0) = v$ and that any other solution will agree with γ on the overlap of their domains. We can package this up into a single vector field on TM which we will call the geodesic field.

Theorem 4.4.5. *Let (M, g) be a Riemannian manifold. Then there is a unique smooth vector field G on TM whose integral curves are of the form $t \mapsto (\gamma(t), \gamma'(t))$, where γ is a geodesic on M .*

Recall from [Definition 1.7.1](#) that an integral curve of a vector field is one whose velocity agrees with the vector field at every point along the curve.

Proof. Suppose such a G exists. Then the trajectories of G must be given by $t \mapsto (\gamma(t), \gamma'(t))$, which must solve the system (4.17) in local coordinates. By uniqueness of the trajectories of such a system, G must be unique.

On the other hand, to prove that such a G exists, we can simply define it in coordinates by the system (4.17). \square

Definition 4.4.6. The vector field $G \in \mathfrak{X}(TM)$ defined in [Theorem 4.4.5](#) is called the *geodesic field* on TM and its local flow is called the *geodesic flow* on TM .

Example 4.4.7. Let $M = S^1$ with its usual metric, namely $g_\theta \left(\frac{\partial}{\partial \theta}, \frac{\partial}{\partial \theta} \right) = 1$. It's straightforward to check using (4.10) that

$$\nabla_{\frac{\partial}{\partial \theta}} \frac{\partial}{\partial \theta} = 0.$$

So then the system (4.17) on $TS^1 = S^1 \times \mathbb{R}$ with coordinates (θ, y) reduces to

$$\begin{aligned}\theta'(t) &= y(t) \\ y'(t) &= 0.\end{aligned}$$

Solutions are of the form $(\gamma(t), \gamma'(t)) = (c_1 + c_2 t, c_2)$, so when we require $\gamma(0) = \theta_0$ and $\gamma'(0) = y_0 \frac{\partial}{\partial \theta} \in T_{\theta_0} S^1$, we get $\gamma(t) = \theta_0 + y_0 t$. These curves are trajectories of the field

$$G(\theta, y) = (y, 0)$$

on the cylinder, as shown in Figure 4.8. This makes sense: geodesics go around the circle at constant speed, and this speed is determined by the initial velocity, which corresponds to height on the cylinder in this picture.

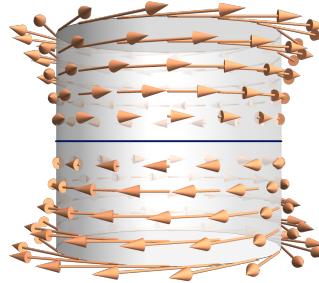


Figure 4.8: The geodesic field on $TS^1 = S^1 \times \mathbb{R}$. The zero section $\{(\theta, 0)\} \cong S^1$ is shown in dark blue.

Example 4.4.8. Let $M = S^2$ with the Riemannian metric induced by the Euclidean metric on \mathbb{R}^3 . We'll work locally in spherical coordinates $(\phi, \theta) \mapsto (\cos \theta \sin \phi, \sin \theta \sin \phi, \cos \phi)$.

Exercise 4.4.9. Show that in these coordinates,

$$\begin{array}{lll}\Gamma_{11}^1 = 0 & \Gamma_{12}^1 = \Gamma_{21}^1 = 0 & \Gamma_{22}^1 = -\sin \phi \cos \phi \\ \Gamma_{11}^2 = 0 & \Gamma_{12}^2 = \Gamma_{21}^2 = \cot \phi & \Gamma_{22}^2 = 0.\end{array}$$

Therefore, in the coordinates (ϕ, θ, y, z) on TS^2 , (4.17) becomes

$$\begin{aligned}\phi' &= y \\ \theta' &= z \\ y' &= (\sin \phi)(\cos \phi)z^2 \\ z' &= -2(\cot \phi)yz.\end{aligned}$$

If we want the geodesic through $(\phi_0, \theta_0) \in S^2$ with initial velocity $v \in T_{(\phi_0, \theta_0)}S^2$, then this gives a well-defined initial value problem.

It's not so clear to me how to solve this problem analytically, but we could certainly solve it numerically for any desired ϕ_0, θ_0, v , and then the velocities of these geodesics would give the geodesic field G on TS^2 (which is a nontrivial bundle: $TS^2 \not\cong S^2 \times \mathbb{R}^2$).

The existence of the geodesic flow follows from [Proposition 1.7.2](#). More precisely, if $U \subset M$ is an open set and $\epsilon > 0$, we'll use the notation

$$TU_\epsilon := \{(p, v) \in TU : |v| < \epsilon\}.$$

Then applying [Proposition 1.7.2](#) to the geodesic field G on TM at the point $(p, 0) \in TM$ yields:

Proposition 4.4.10. *Let M be a Riemannian manifold and let $p \in M$. There is an open set U containing p , real numbers $\delta, \epsilon > 0$, and a smooth map $\phi: (-\delta, \delta) \times TU_\epsilon \rightarrow TU$ so that $t \mapsto \phi(t, q, v)$ is the unique trajectory of the geodesic field G with $\phi(0, q, v) = (q, v)$ for all $(q, v) \in TU_\epsilon$.*

Composing the local flow ϕ with the projection $\pi: TM \rightarrow M$ yields the map $\gamma = \pi \circ \phi$ which satisfies:

Proposition 4.4.11. *Let M be a Riemannian manifold and let $p \in M$. There is an open set U containing p , real numbers $\delta, \epsilon > 0$, and a smooth map $\gamma: (-\delta, \delta) \times TU_\epsilon \rightarrow M$ so that $t \mapsto \gamma(t, q, v)$ is the unique geodesic in M passing through q at time 0 with instantaneous velocity v for any $(q, v) \in TU_\epsilon$.*

Notice that we can always make δ bigger by making ϵ smaller (or *vice versa*), so we can get δ of uniform size if we want:

Corollary 4.4.12. *Let M be a Riemannian manifold and let $p \in M$. There is an open set U containing p , a real number $\epsilon > 0$, and a smooth map $\gamma: (-2, 2) \times TU_\epsilon \rightarrow M$ so that $t \mapsto \gamma(t, q, v)$ is the unique geodesic in M passing through q at time 0 with instantaneous velocity v for any $(q, v) \in TU_\epsilon$.*

In particular, this means that, so long as v is short enough, $\gamma(1, q, v)$ is well-defined. We should interpret $\gamma(1, q, v)$ as follows: we travel along the geodesic through q in the direction of the unit vector $\frac{v}{\|v\|}$ for a distance of $\|v\|$. This idea of going out in the manifold in the direction indicated by v for a distance $\|v\|$ is so useful that we give it a special name:

Definition 4.4.13. The map $\exp: TU_\epsilon \rightarrow M$ defined by

$$\exp(q, v) = \gamma(1, q, v)$$

is called the (Riemannian) *exponential map*.

Most of the time, we will just be interested in the restriction of \exp to the tangent space at a point. That is, if $B_\epsilon(0) \subset T_q M$ is the open ball of radius ϵ centered at the origin in $T_q M$, then we will define

$$\exp_q: B_\epsilon(0) \rightarrow M$$

by $\exp_q(v) = \exp(q, v)$.

In the case that M is *geodesically complete*, meaning that every geodesic can be extended forever in both directions, then we can extend this map to the entire tangent space:

$$\exp_q: T_q M \rightarrow M.$$

Theorem 4.4.14. Given $q \in M$, there is some $\epsilon > 0$ so that $\exp_q : B_\epsilon(0) \rightarrow M$ is a diffeomorphism of $B_\epsilon(0) \subset T_q M$ onto an open neighborhood of q in M .

Proof. We can compute the differential of \exp at the origin:

$$\begin{aligned} d(\exp_q)_0(v) &= \frac{d}{dt} \Big|_{t=0} \exp_q(tv) \\ &= \frac{d}{dt} \Big|_{t=0} \gamma(1, q, tv) \\ &= \frac{d}{dt} \Big|_{t=0} \gamma(t, q, v) \\ &= v, \end{aligned}$$

by definition of $\gamma(t, q, v)$.

In other words, $d(\exp_q)_0$ is the identity map on $T_q M$, so the inverse function theorem (Theorem 1.4.7) implies that \exp_q is a local diffeomorphism. \square

4.5 Geodesics on Lie Groups

In Section 4.4 we defined geodesics as the curves $\gamma : I \rightarrow M$ with $\frac{D\gamma'}{dt} = 0$, which means that geodesics are parametrized at constant speed and are straight (in the sense that they don't turn). Compare to the fact that straight lines $\alpha : I \rightarrow \mathbb{R}^n$ traversed at constant speed satisfy $\alpha''(t) = 0$.

It seems intuitively obvious that the shortest path between two points should be as straight as possible, and hence should be a geodesic. Of course, we have to be careful because there is generally not a *unique* geodesic between two points (think about points on the sphere) and there may be no geodesic at all (think about the points $(2, 0)$ and $(-2, 0)$ in the annulus $\{(x, y) \in \mathbb{R}^2 : 1 < x^2 + y^2 < 4\}$). But if two points are close enough, there will be a unique minimizing geodesic between them.

But, from Theorem 4.4.14, we know that for any $p \in M$ there exists $\epsilon > 0$ so that $\exp_p : B_\epsilon(0) \rightarrow M$ is a diffeomorphism. The image $\exp_p(B_\epsilon(0))$ is called a *normal ball* or *geodesic ball* of radius ϵ around p , and denoted $B_\epsilon(p)$.

Theorem 4.5.1. Within any normal ball $B_\epsilon(p)$, the geodesic rays from p are the unique shortest paths from p to their endpoints.

The proof of this theorem is fairly technical and not very enlightening, so we skip it. For details, see do Carmo [7, §3.3].

Instead, let's go a bit deeper on geodesics on Lie groups.

4.5.1 Bi-Invariant Metrics

First of all, recall from Example 4.1.6 that we can define a left-invariant Riemannian metric on any Lie group G by choosing an inner product $\langle \cdot, \cdot \rangle$ on $T_e G \cong \mathfrak{g}$, and then pulling it back to each tangent space: for $h \in G$ and $u, v \in T_h G$,

$$g_h(u, v) := \langle (dL_{h^{-1}})_h u, (dL_{h^{-1}})_h v \rangle_e, \quad (4.18)$$

and similarly we could have gotten a right-invariant metric using $R_{h^{-1}}$ rather than $L_{h^{-1}}$.

Formally, we define a *left-invariant Riemannian metric* on G by requiring that

$$g_{h_1}(u, v) = g_{h_2 h_1}((dL_{h_2})_{h_1} u, (dL_{h_2})_{h_1} v)$$

for each $h_1, h_2 \in G$ and each $u, v \in T_{h_1} G$. Note that $(dL_{h_2})_{h_1} : T_{h_1} G \rightarrow T_{L_{h_2}(h_1)} G = T_{h_2 h_1} G$, so the right-hand side makes sense.

In other words, a metric is left-invariant if all left translations are isometries of G .

Exercise 4.5.2. Show that the metric g defined by (4.18) is left-invariant in the above sense.

Similarly, we say that a metric on G is *right-invariant* if all right translations are isometries.

In general, there are lots of left-invariant metrics on a Lie group, and also lots of right-invariant metrics. And it's not so clear which one to pick, or even whether to pick a left-invariant one or a right-invariant one.

However, in certain special cases we get Riemannian metrics on Lie groups which are both left-invariant and right-invariant. We call such metrics *bi-invariant* and they are, in some sense, the best possible Riemannian metrics to put on Lie groups: they are highly symmetric, and we don't have to choose whether we prefer left- or right-invariance. Moreover, under appropriate hypotheses, bi-invariant metrics are essentially unique; see [Theorem 4.5.9](#).

Note that left/right/bi-invariant metrics on G are completely determined by the inner product

$$\langle \cdot, \cdot \rangle_e = g_e(\cdot, \cdot)$$

on $T_e G \cong \mathfrak{g}$; after all, any metric on G determines such an inner product, and such an inner product uniquely determines a left-invariant metric using (4.18) (or a right-invariant metric using the analogous formula replacing $L_{h^{-1}}$ with $R_{h^{-1}}$; if the two agree, then what either gives is a bi-invariant metric).

Proposition 4.5.3. *Suppose the Lie group G admits a bi-invariant metric. Then the induced inner product on \mathfrak{g} satisfies*

$$\langle [X, Y], Z \rangle_e = \langle Y, [Z, X] \rangle_e$$

for all $X, Y, Z \in \mathfrak{g}$.

Proof. Since $X, Y, Z \in \mathfrak{g}$, they are (or can be identified with) left-invariant vector fields on G . Let Φ_t be the local flow associated to X . Because the metric is bi-invariant we know that

$$\begin{aligned} \langle Y, Z \rangle_e &= g_e(Y, Z) = g_e\left((dR_{\Phi_t(e)})_{\Phi_t^{-1}(e)}(dL_{\Phi_t^{-1}(e)})_e Y_e, (dR_{\Phi_t(e)})_{\Phi_t^{-1}(e)}(dL_{\Phi_t^{-1}(e)})_e Z_e\right) \\ &= g_e\left((dR_{\Phi_t(e)})_{\Phi_t^{-1}(e)} Y_{\Phi_t^{-1}(e)}, (dR_{\Phi_t(e)})_{\Phi_t^{-1}(e)} Z_{\Phi_t^{-1}(e)}\right) \\ &= \langle (dR_{\Phi_t(e)})_{\Phi_t^{-1}(e)} Y_{\Phi_t^{-1}(e)}, (dR_{\Phi_t(e)})_{\Phi_t^{-1}(e)} Z_{\Phi_t^{-1}(e)} \rangle_e \end{aligned}$$

by left-invariance of Y and Z .

Differentiating the above expression with respect to t , evaluating at $t = 0$, and recalling that the Lie bracket is the same as the Lie derivative eventually yields

$$0 = \langle [Y, X], Z \rangle_e + \langle Y, [Z, X] \rangle,$$

which is equivalent to the stated equation. □

If we set $X = Z$ in Proposition 4.5.3, then we see that

$$\langle [X, Y], X \rangle_e = 0.$$

In other words:

Corollary 4.5.4. *In a bi-invariant metric, the bracket of two left-invariant vector fields X and Y is orthogonal to both X and Y .*

In fact, the converse of Proposition 4.5.3 is also true, so we get the following characterization of bi-invariant metrics on Lie groups:

Theorem 4.5.5. *A Riemannian metric on a Lie group G is bi-invariant if and only if the induced inner product on \mathfrak{g} satisfies*

$$\langle [X, Y], Z \rangle_e = \langle Y, [Z, X] \rangle_e$$

for all $X, Y, Z \in \mathfrak{g}$.

Example 4.5.6. The goal here is to show that $\text{Aff}^+(\mathbb{R})$ does *not* admit a bi-invariant metric. Recall from Exercise 3.3.10 that we have left-invariant vector fields $A, B \in \mathfrak{aff}^+(\mathbb{R})$ with $[A, B] = B$. If $\text{Aff}^+(\mathbb{R})$ admits a bi-invariant metric, then Proposition 4.5.3 (or Theorem 4.5.5) implies that the induced inner product on $\mathfrak{aff}^+(\mathbb{R}) \cong \mathbb{R}^2$ satisfies

$$0 = \langle [B, B], A \rangle_e = \langle B, [B, A] \rangle_e = \langle B, -B \rangle_e = -\langle B, B \rangle_e \neq 0$$

since B is not the zero vector and the inner product must be positive-definite. So there is no such bi-invariant metric.

It is a bit unfortunate that not all Lie groups admit bi-invariant metrics, but it turns out that all *compact* Lie groups do. In service of proving this, let's first record the following observation:

Lemma 4.5.7. *A Lie group G admits a bi-invariant Riemannian metric if and only if there is an inner product on $T_e G$ which is invariant under conjugation.*

Proof. (\Rightarrow) For the forward direction, suppose G has a bi-invariant metric g . Let $\langle \cdot, \cdot \rangle_e$ be the induced inner product on $T_e G$, and observe that, for any $h \in G$ and any $X, Y \in \mathfrak{g}$,

$$\begin{aligned} \langle (dC_h)_e X_e, (dC_h)_e Y_e \rangle_e &= \langle d(L_h \circ R_{h^{-1}})_e X_e, d(L_h \circ R_{h^{-1}})_e Y_e \rangle_e = \langle (dL_h)_{h^{-1}}(dR_{h^{-1}})_e X_e, (dL_h)_{h^{-1}}(dR_{h^{-1}})_e Y_e \rangle_e \\ &= g_{h^{-1}}((dR_{h^{-1}})_e X_e, (dR_{h^{-1}})_e Y_e) = g_e(X_e, Y_e) = \langle X_e, Y_e \rangle_e \end{aligned}$$

by left- and right-invariance of g . So $\langle \cdot, \cdot \rangle_e$ is conjugation-invariant.

(\Leftarrow) On the other hand, suppose $\langle \cdot, \cdot \rangle_e$ is conjugation-invariant. Define the left-invariant metric g by (4.18). I claim that g is also right-invariant, and hence bi-invariant. To see this, let $h_1, h_2 \in G$. Notice that, for any $k \in G$,

$$(L_{(h_1 h_2)^{-1}} \circ R_{h_2})(k) = h_2^{-1} h_1^{-1} k h_2 = C_{h_2^{-1}}(L_{h_1^{-1}}(k)) = (C_{h_2^{-1}} \circ L_{h_1^{-1}})(k),$$

and hence $L_{(h_1 h_2)^{-1}} \circ R_{h_1} = C_{h_2^{-1}} \circ L_{h_1^{-1}}$.

Therefore, for any vector fields $u, v \in T_{h_1}G$,

$$\begin{aligned} g_{h_1 h_2}((dR_{h_2})_{h_1} u, (dR_{h_2})_{h_1} v) &= \langle (dL_{(h_1 h_2)^{-1}})_{h_1 h_2} (dR_{h_2})_{h_1} u, (dL_{(h_1 h_2)^{-1}})_{h_1 h_2} (dR_{h_2})_{h_1} v \rangle_e \\ &= \langle (d(L_{(h_1 h_2)^{-1}} \circ R_{h_2}))_{h_1} u, (d(L_{(h_1 h_2)^{-1}} \circ R_{h_2}))_{h_1} v \rangle_e \\ &= \langle (d(C_{h_2^{-1}} \circ L_{h_1^{-1}}))_{h_1} u, (d(C_{h_2^{-1}} \circ L_{h_1^{-1}}))_{h_1} v \rangle_e \\ &= \langle (dC_{h_2^{-1}})_e (dL_{h_1^{-1}})_{h_1} u, (dC_{h_2^{-1}})_e (dL_{h_1^{-1}})_{h_1} v \rangle_e \end{aligned}$$

by the chain rule. But now conjugation-invariance of $\langle \cdot, \cdot \rangle_e$ implies this is equal to

$$\langle (dL_{h_1^{-1}})_{h_1} u, (dL_{h_1^{-1}})_{h_1} v \rangle_e = g_{h_1}(u, v)$$

by definition of g . Stringing these equalities together gives

$$g_{h_1 h_2}((dR_{h_2})_{h_1} u, (dR_{h_2})_{h_1} v) = g_{h_1}(u, v),$$

which is the definition of right-invariance. So we see that the left-invariant metric g is also right-invariant, and hence it is bi-invariant. \square

We're now ready to show that all *compact* Lie groups admit a bi-invariant metric. The strategy is to pick any arbitrary inner product on $T_e G$ and then average it over all conjugations. If the group is non-compact, then this could easily fail: the averaging procedure might fail to converge. But when G is compact, this isn't a problem:

Theorem 4.5.8. *Every compact Lie group G admits a bi-invariant Riemannian metric.*

Proof. Start with a right-invariant metric on G , and let $d\text{Vol} \in \Omega^n(G)$ be the associated right-invariant volume form guaranteed by Definition 4.1.11. That is, $d\text{Vol}(h) \in \wedge^n(T_h G)^*$ for all $h \in G$.

Now, let $\langle \cdot, \cdot \rangle_e$ be an arbitrary inner product on $T_e G$. Then we define a new inner product $\langle \cdot, \cdot \rangle'_e$ by averaging over conjugations:

$$\langle u, v \rangle'_e := \int_{h \in G} \langle (dC_h)_e u, (dC_h)_e v \rangle_e d\text{Vol}(h).$$

Here's where we using compactness, which guarantees that this integral converges. Right-invariance of the volume form now implies that

$$\begin{aligned} \langle (dC_{h_1})_e u, (dC_{h_1})_e v \rangle'_e &= \int_G \langle (dC_{h_2})_e (dC_{h_1})_e u, (dC_{h_2})_e (dC_{h_1})_e v \rangle_e d\text{Vol}(h_2) \\ &= \int_G \langle (d(C_{h_2} \circ C_{h_1}))_e u, (d(C_{h_2} \circ C_{h_1}))_e v \rangle_e d\text{Vol}(h_2) \\ &= \int_G \langle (d(C_{h_2 h_1}))_e u, (d(C_{h_2 h_1}))_e v \rangle_e d\text{Vol}(h_2) \\ &= \int_G \langle (d(C_{h_2 h_1}))_e u, (d(C_{h_2 h_1}))_e v \rangle_e R_{h_1}^* d\text{Vol}(h_2) \\ &= \int_G \langle (d(C_{h_2 h_1}))_e u, (d(C_{h_2 h_1}))_e v \rangle_e d\text{Vol}(h_2 h_1) \\ &= \int_G \langle (d(C_h))_e u, (d(C_h))_e v \rangle_e d\text{Vol}(h) \\ &= \langle u, v \rangle'_e \end{aligned}$$

using the change of variables formula for the map $R_{h_1}: G \rightarrow G$, the right-invariance of $d\text{Vol}$, and then making the substitution $h = h_2h_1$.

Therefore, $\langle \cdot, \cdot \rangle_e$ is conjugation-invariant, and hence corresponds to a bi-invariant metric on G by Lemma 4.5.7. \square

When G is simple, the bi-invariant metric we get is essentially unique:

Theorem 4.5.9 (see, e.g., [1, Proposition 2.48]). *Let G be a compact, simple Lie group. Then the bi-invariant metric on G is unique up to re-scaling.*

4.5.2 Geodesics in the Bi-Invariant Metric

Recall that we defined an exponential map $\exp: \mathfrak{g} \cong T_e G \rightarrow G$ for Lie groups in Section 3.10. We also defined a *Riemannian* exponential map $\exp_e: T_e G \rightarrow G$ in Definition 4.4.13. Of course, the Riemannian exponential depends on a choice of Riemannian metric, whereas the Lie group exponential map only depends on the Lie group structure. However, when we have a bi-invariant metric, the two exponential maps will agree, which is the main point of this section.

Example 4.5.10. Recall our left-invariant metric on $\text{Aff}^+(\mathbb{R}) \cong H$ from Example 4.1.7. We saw in Example 4.4.4 that the geodesics for this metric are vertical lines and semi-circles centered at points on the x -axis. Therefore, the geodesics through the identity $(0, 1)$ are as depicted in Figure 4.9.

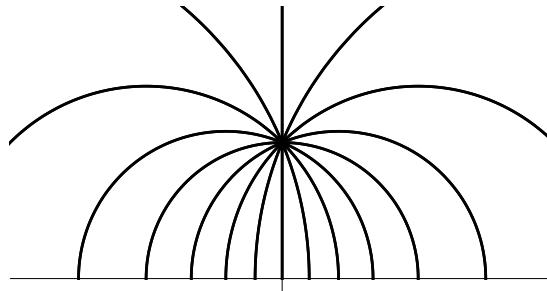


Figure 4.9: The geodesics through the identity in $\text{Aff}^+(\mathbb{R}) \cong H$.

On the other hand, suppose $v = a \frac{\partial}{\partial x} + b \frac{\partial}{\partial y} \in T_{(0,1)} H$ is a tangent vector at the identity. Under the identification of $\text{Aff}^+(\mathbb{R})$ with the matrix group $\left\{ \begin{bmatrix} y & x \\ 0 & 1 \end{bmatrix} : (x, y) \in \mathbb{R}^2, y > 0 \right\}$ from Example 3.2.3, v is identified with the 2×2 matrix

$$A_v = \begin{bmatrix} b & a \\ 0 & 0 \end{bmatrix} \in T_I \text{GL}_2(\mathbb{R}).$$

In this setting, the exponential map is just the matrix exponential, so

$$\exp(tv) = I + tA_v + \frac{1}{2!}(tA_v)^2 + \dots = \begin{bmatrix} 1 & 0 \\ 0 & 1 \end{bmatrix} + t \begin{bmatrix} b & a \\ 0 & 0 \end{bmatrix} + \frac{t^2}{2!} \begin{bmatrix} b^2 & ab \\ 0 & 0 \end{bmatrix} + \frac{t^3}{3!} \begin{bmatrix} b^3 & ab^2 \\ 0 & 0 \end{bmatrix} + \dots = \begin{bmatrix} e^{tb} & \frac{a(e^{tb}-1)}{b} \\ 0 & 1 \end{bmatrix}.$$

In other words, the one-parameter subgroup associated to $v \in T_{(0,1)} H$ is the curve $t \mapsto \left(\frac{a(e^{tb}-1)}{b}, e^{tb} \right)$, as depicted in Figure 4.10.

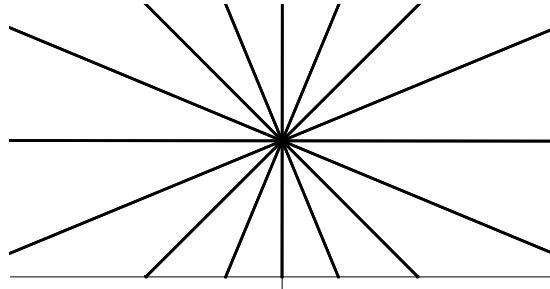


Figure 4.10: The one-parameter subgroups in H .

Clearly, the one-parameter subgroups are *not* geodesics with respect to the metric. The issue is, as we've already seen in [Example 4.5.6](#), that $\text{Aff}^+(\mathbb{R})$ does not have a bi-invariant metric.

Theorem 4.5.11. *In a bi-invariant metric on a Lie group, the one-parameter subgroups coincide with the geodesics through the identity.*

Proof. Recall from [\(4.10\)](#) the formula

$$g(\nabla_V U, W) = \frac{1}{2} [U(g(V, W)) + V(g(W, U)) - W(g(U, V)) - g([U, W], V) - g([V, W], U) - g([U, V], W)],$$

Assume $V, W \in \mathfrak{aff}^+(\mathbb{R})$ are left-invariant, and let $U = V$ in the above equation. Since the metric is bi-invariant, $g(V, W)$ and $g(V, V)$ are constant, so the first three terms all vanish. The last term also vanishes since $[V, V] = 0$, so we're left with

$$g(\nabla_V V, W) = \frac{1}{2} [-g([V, W], V) - g([V, W], V)] = -g([V, W], V) = g([W, V], V) = 0$$

by [Corollary 4.5.4](#).

Since W could have been any left-invariant vector field, this implies that $\nabla_V V = 0$ for all left-invariant vector fields V , where ∇ is the Levi-Civita connection. But then if $\gamma(t) = \exp(tV_e)$ is the one-parameter subgroup associated with $V_e \in T_e G$, we have that

$$\gamma'(t) = \frac{d}{dt} \exp(tV_e) = (dL_{\exp(tV_e)})_e V_e = V_{\exp(tV_e)},$$

so V is precisely the velocity field of the curve γ , and hence

$$0 = \nabla_V V = \nabla_{\gamma'(t)} \gamma'(t) = \frac{D\gamma'}{dt}$$

implies that γ is a geodesic. □

Since the Lie group exponential is defined in terms of one-parameter subgroups and the Riemannian exponential is defined in terms of geodesics, [Theorem 4.5.11](#) implies that the Lie group exponential and the Riemannian exponential are the same on a Lie group with a bi-invariant metric.

In particular, this means that on matrix groups with a bi-invariant metric we can often solve the following problem: given $h_1, h_2 \in G$, what is the minimizing geodesic from h_1 to h_2 . To do so, note that it suffices to find a geodesic $\gamma(t)$ from the identity element e to $h_1^{-1}h_2$: since left-multiplication is an isometry, $\alpha(t) = h_1\gamma(t)$ will be a geodesic from h_1 to h_2 .

So we've reduced to finding the geodesic from the identity to any group element h . By [Theorem 4.5.11](#), this geodesic will be a one-parameter subgroup passing through h , so our goal is to find, say, a unit vector $v \in T_e G$ so that $\exp(t_0 v) = h$. Then the curve $\gamma: [0, t_0] \rightarrow G$ given by $\gamma(t) = \exp(tv)$ will be such a geodesic, and it will have length t_0 .

But inverting the matrix exponential is exactly what the matrix logarithm does. In general, since the log is multivalued, $\log(h)$ will be a collection of elements of $T_e G$, but if we choose u with smallest norm and let $v = \frac{u}{\|u\|}$ then it will follow that $d(e, h) = \|u\|$ and $\gamma(t) = \exp(tv)$ is a minimizing geodesic from e to h .

Example 4.5.12. Consider $G = \mathrm{SO}(3)$. If $A \in \mathrm{SO}(3)$, then A is unitarily diagonalizable:

$$A = UDU^*,$$

where $U \in \mathrm{U}(3)$ and $D = \mathrm{diag}(1, e^{i\theta}, e^{-i\theta})$ is the diagonal matrix with eigenvalues of A on the diagonal. Then

$$\log(A) = U \log(D) U^* = U \mathrm{diag}(0, i\theta, -i\theta) U^*.$$

So then the distance from I to A is

$$\|\log(A)\| = \sqrt{2}\theta$$

and the unit-speed geodesic from I to A is

$$\gamma(t) = U \mathrm{diag}(1, e^{it/\sqrt{2}}, e^{-it/\sqrt{2}}) U^*,$$

with $A = \gamma(\sqrt{2}\theta)$.

Of course, the same sort of thing works on $\mathrm{SO}(n)$ for any n .

4.6 Geodesics on Grassmannians

The goal in this section is to describe geodesics on Grassmannians. This is knowledge that I have used many times in my research career; since Grassmannians pop up everywhere from string theory to random polygons to hyperspectral imaging to video analysis to algebraic geometry, you may find this information important at some point, too. The information in this section is mostly drawn from Edelman, Arias, and Smith's amazing paper [\[9\]](#).

Recall from [Section 3.12](#) that the Grassmannian $\mathrm{Gr}(k, \mathbb{R}^n)$ is the collection of k -dimensional linear subspaces of \mathbb{R}^n , and that it can be realized as a quotient of $\mathrm{O}(n)$ by

$$\mathrm{Gr}(k, \mathbb{R}^n) \cong \mathrm{O}(n)/(\mathrm{O}(k) \times \mathrm{O}(n-k)).$$

In other words, we can interpret points in the Grassmannian as equivalence classes $[Q]$ of all orthogonal matrices whose first k columns span the same subspace as those of $Q \in \mathrm{O}(n)$. Concretely,

$$[Q] := \left\{ Q \begin{bmatrix} U_k & 0 \\ 0 & U_{n-k} \end{bmatrix} : U_k \in \mathrm{O}(k), U_{n-k} \in \mathrm{O}(n-k) \right\}.$$

The tangent space $T_Q \mathrm{O}(n)$ splits into *horizontal* and *vertical* components: the vertical space is the subspace tangent to $[Q]$ (i.e., moving in those directions does not change the point on the Grassmannian), and the horizontal space is the orthogonal complement of the vertical space.

From the above description of $[Q]$, it should be pretty clear that the vertical space is

$$\left\{ Q \begin{bmatrix} A & 0 \\ 0 & C \end{bmatrix} : A \text{ is } k \times k \text{ skew-symmetric and } C \text{ is } (n-k) \times (n-k) \text{ skew-symmetric} \right\}.$$

Hence, the horizontal space, which we can identify with $T_{[Q]} \mathrm{Gr}(k, \mathbb{R}^n)$, consists of matrices of the form

$$\Delta = Q \begin{bmatrix} 0 & -B^T \\ B & 0 \end{bmatrix},$$

where B is an arbitrary $(n-k) \times k$ matrix. Note that this immediately recovers the fact that $\dim(\mathrm{Gr}(k, \mathbb{R}^n)) = k(n-k)$.

We get the standard Riemannian metric on $\mathrm{Gr}(k, \mathbb{R}^n)$ by restricting the metric on $\mathrm{O}(n)$ to the horizontal space (it is also traditional to multiply by $1/2$ because of the repeated B 's in the expression for Δ above). So what is the metric on the orthogonal group?

We get a left-invariant metric on $\mathrm{O}(n)$ by pushing around any inner product we like on $T_I \mathrm{O}(n) \subset T_I \mathrm{GL}_n(\mathbb{R}) = \mathrm{Mat}_{n \times n}(\mathbb{R})$, where the standard inner product is the Frobenius inner product

$$\langle M_1, M_2 \rangle_{\mathrm{Fr}} = \mathrm{tr}(M_1^T M_2).$$

When we restrict this to $T_I \mathrm{O}(n)$, which consists of the skew-symmetric $n \times n$ matrices, we see that

$$\begin{aligned} \langle [X, Y], Z \rangle_{\mathrm{Fr}} &= \mathrm{tr}([X, Y]^T Z) = \mathrm{tr}((XY - YX)^T Z) = \mathrm{tr}(Y^T X^T Z - X^T Y^T Z) = \mathrm{tr}(Y^T X^T Z - Y^T ZX^T) \\ &= \mathrm{tr}(-Y^T XZ + Y^T ZX) = \mathrm{tr}(Y^T [Z, X]) = \langle Y, [Z, X] \rangle_{\mathrm{Fr}} \end{aligned}$$

using cyclic invariance of the trace and the fact that $X^T = -X$. Therefore, [Theorem 4.5.5](#) implies that the left-invariant metric $g_{\mathrm{O}(n)}$ on $\mathrm{O}(n)$ induced by $\langle \cdot, \cdot \rangle_{\mathrm{Fr}}$ is bi-invariant.

What is this metric? First, recall that, for $Q \in \mathrm{O}(n)$, $T_Q \mathrm{O}(n) = \{QX : X \text{ is skew-symmetric}\}$ since $T_I \mathrm{O}(n)$ consists of skew-symmetric matrices and $(dL_Q)_I X = QX$. Then, for $QX, QY \in T_Q \mathrm{O}(n)$,

$$g_Q^{\mathrm{O}(n)}(QX, QY) = \langle (dL_{Q^{-1}})_Q QX, dL_{Q^{-1}})_Q QY \rangle_{\mathrm{Fr}} = \langle Q^{-1}QX, Q^{-1}QY \rangle_{\mathrm{Fr}} = \langle X, Y \rangle_{\mathrm{Fr}} = \mathrm{tr}(X^T Y).$$

Note that this is really just the Frobenius inner product of QX with QY :

$$\langle QX, QY \rangle_{\mathrm{Fr}} = \mathrm{tr}((QX)^T (QY)) = \mathrm{tr}(X^T Q^T QY) = \mathrm{tr}(X^T Y).$$

Therefore, with the factor of $1/2$ mentioned previously, the induced metric on $\mathrm{Gr}(k, \mathbb{R}^n)$ is given by, for $\Delta_i = Q \begin{bmatrix} 0 & -B_i^T \\ B_i & 0 \end{bmatrix} \in T_{[Q]} \mathrm{Gr}(k, \mathbb{R}^n)$,

$$g_{[Q]}(\Delta_1, \Delta_2) = \frac{1}{2} \mathrm{tr} \left(\begin{bmatrix} 0 & B_1^T \\ -B_1 & 0 \end{bmatrix} \begin{bmatrix} 0 & -B_2^T \\ B_2 & 0 \end{bmatrix} \right) = \frac{1}{2} \mathrm{tr}(B_1^T B_2 + B_1 B_2^T) = \mathrm{tr}(B_1^T B_2)$$

From [Theorem 4.5.11](#) we know that geodesics on $\mathrm{O}(n)$ take the form $\gamma(t) = Qe^{tX}$ where $Q \in \mathrm{O}(n)$, X is skew-symmetric, and we have $\gamma(0) = Q$ and $\gamma'(0) = QX \in T_Q \mathrm{O}(n)$.

If $Q \in O(n)$ and $\Delta = Q \begin{bmatrix} 0 & -B^T \\ B & 0 \end{bmatrix} \in T_{[Q]} \text{Gr}(k, \mathbb{R}^n)$, then

$$\gamma(t) = Q e^{t \begin{bmatrix} 0 & -B^T \\ B & 0 \end{bmatrix}}$$

is a horizontal curve: the velocity vector $\gamma'(t) = Q e^{t \begin{bmatrix} 0 & -B^T \\ B & 0 \end{bmatrix}} \begin{bmatrix} 0 & -B^T \\ B & 0 \end{bmatrix}$ is in the horizontal space at $\gamma(t) = Q e^{t \begin{bmatrix} 0 & -B^T \\ B & 0 \end{bmatrix}}$. This means that $[\gamma(t)]$ is a geodesic in the Grassmannian, and conversely it can be shown that every geodesic in $\text{Gr}(k, \mathbb{R}^n)$ has a horizontal lift of this form.

In general, when we're working with the Grassmannian, we'd rather avoid working with $n \times n$ matrices (especially when k is much smaller than n). Since we also know that $\text{Gr}(k, \mathbb{R}^n) = \text{St}(k, \mathbb{R}^n)/O(k)$, we can also represent points in the Grassmannian by equivalence classes of $n \times k$ matrices Y with orthonormal columns:

$$[Y] = \{YU_k : U_k \in O(k)\}.$$

In practice, we can just think of Y as the first k columns of the orthogonal matrix Q from before. The above expression makes it clear that the vertical space of the projection $\text{St}(k, \mathbb{R}^n) \rightarrow \text{Gr}(k, \mathbb{R}^n) = \text{St}(k, \mathbb{R}^n)/O(k)$ consists of $n \times k$ matrices of the form

$$YA,$$

where A is $k \times k$ skew-symmetric; that is, it corresponds to infinitesimal rotations of the basis for the subspace given by the columns of Y . Therefore, the horizontal space, which we can identify with $T_{[Y]} \text{Gr}(k, \mathbb{R}^n)$, consists of all $n \times k$ matrices H whose columns are perpendicular to the column space of Y ; that is,

$$Y^T H = 0.$$

Here's another way to see this: if Y consists of the first k columns of an orthogonal matrix $Q = [Y \ Z]$, then the first k columns of

$$Q \begin{bmatrix} 0 & -B^T \\ B & 0 \end{bmatrix} = [Y \ Z] \begin{bmatrix} 0 & -B^T \\ B & 0 \end{bmatrix} = [ZB \ -YB^T]$$

are $H = ZB$, and, since the columns of Y are perpendicular to the columns of Z ,

$$Y^T H = Y^T ZB = 0.$$

In terms of $n \times k$ matrices, the Riemannian metric on the Grassmannian is still just the Frobenius inner product: if $\Delta_i = Q \begin{bmatrix} 0 & -B_i^T \\ B_i & 0 \end{bmatrix}$ and $H_i = ZB_i$ is the $n \times k$ matrix consisting of the first k columns of Δ_i , then

$$\text{tr}(H_1^T H_2) = \text{tr}(B_1^T Z^T ZB_2) = \text{tr}(B_1^T B_2) = g_{[Q]}(\Delta_1, \Delta_2).$$

If we're working with $n \times k$ matrices, we would then write

$$g_{[Y]}(H_1, H_2) = \text{tr}(H_1^T H_2).$$

We can write geodesics in $n \times k$ matrix format as well by just chopping off the last $n - k$ columns; now our geodesics look like

$$\gamma(t) = Q e^{t \begin{bmatrix} 0 & -B^T \\ B & 0 \end{bmatrix}} I_{n,k},$$

where $I_{n,k}$ consists of the first k columns of the $n \times n$ identity matrix. Of course, it would be preferable to write this in terms of the $n \times k$ matrices Y and H instead of the $n \times n$ matrices Q and $\begin{bmatrix} 0 & -B^T \\ B & 0 \end{bmatrix}$.

Theorem 4.6.1. Let $\gamma(t) = Q e^{t \begin{bmatrix} 0 & -B^T \\ B & 0 \end{bmatrix}} I_{n,k}$ be the geodesic with $\gamma(0) = Y$ and $\gamma'(0) = H$. Then

$$\gamma(t) = [YV \quad U] \begin{bmatrix} \cos(t\Sigma) \\ \sin(t\Sigma) \end{bmatrix} V^T,$$

where $U\Sigma V^T$ is the singular value decomposition of H .

Proof. Let

$$B = [U_1 \quad U_2] \begin{bmatrix} \Sigma \\ 0 \end{bmatrix} V^T$$

be the singular value decomposition of B . Then

$$\begin{bmatrix} V & 0 & 0 \\ 0 & U_1 & U_2 \\ 0 & 0 & 0 \end{bmatrix} \begin{bmatrix} 0 & -\Sigma & 0 \\ \Sigma & 0 & 0 \\ 0 & 0 & 0 \end{bmatrix} \begin{bmatrix} V^T & 0 \\ 0 & U_1^T \\ 0 & U_2^T \end{bmatrix} = \begin{bmatrix} 0 & -V\Sigma U_1^T \\ U_1 \Sigma V^T & 0 \end{bmatrix} = \begin{bmatrix} 0 & -B^T \\ B & 0 \end{bmatrix}.$$

Therefore,

$$\begin{aligned} e^{t \begin{bmatrix} 0 & -B^T \\ B & 0 \end{bmatrix}} &= e^{t \begin{bmatrix} V & 0 & 0 \\ 0 & U_1 & U_2 \\ 0 & 0 & 0 \end{bmatrix} \begin{bmatrix} 0 & -\Sigma & 0 \\ \Sigma & 0 & 0 \\ 0 & 0 & 0 \end{bmatrix} \begin{bmatrix} V^T & 0 \\ 0 & U_1^T \\ 0 & U_2^T \end{bmatrix}} = \begin{bmatrix} V & 0 & 0 \\ 0 & U_1 & U_2 \\ 0 & 0 & 0 \end{bmatrix} e^{t \begin{bmatrix} 0 & -\Sigma & 0 \\ \Sigma & 0 & 0 \\ 0 & 0 & 0 \end{bmatrix}} \begin{bmatrix} V^T & 0 \\ 0 & U_1^T \\ 0 & U_2^T \end{bmatrix} \\ &= \begin{bmatrix} V & 0 & 0 \\ 0 & U_1 & U_2 \\ 0 & 0 & 0 \end{bmatrix} \begin{bmatrix} \cos(t\Sigma) & -\sin(t\Sigma) & 0 \\ \sin(t\Sigma) & \cos(t\Sigma) & 0 \\ 0 & 0 & I \end{bmatrix} \begin{bmatrix} V^T & 0 \\ 0 & U_1^T \\ 0 & U_2^T \end{bmatrix} \end{aligned} \quad (4.19)$$

and hence (with some intermediate computations omitted),

$$\gamma(t) = Q e^{t \begin{bmatrix} 0 & -B^T \\ B & 0 \end{bmatrix}} I_{n,k} = [YV \quad ZU_1] \begin{bmatrix} \cos(t\Sigma) \\ \sin(t\Sigma) \end{bmatrix} V^T.$$

But now $H = ZB = (ZU_1)\Sigma V^T$ is the singular value decomposition of H , so the result follows. \square

If $H \in T_{[Y]} \text{Gr}(k, \mathbb{R}^n)$ then, at least for small t , Theorem 4.5.1 implies that the distance from $\gamma(0) = Y$ to $\gamma(t)$ is

$$d(\gamma(0), \gamma(t)) = t \|H\|_{\text{Fr}} = t \sqrt{\sum_{i=1}^k \sigma_i^2},$$

where the σ_i are the singular values of H ; that is, the diagonal elements of Σ .

If we write $\theta_i = t\sigma_i$ and let $\Theta = (\theta_1, \dots, \theta_k)$, then (4.19) becomes

$$e^{t \begin{bmatrix} 0 & -B^T \\ B & 0 \end{bmatrix}} = \begin{bmatrix} V & 0 & 0 \\ 0 & U_1 & U_2 \\ 0 & 0 & 0 \end{bmatrix} \begin{bmatrix} \cos \Theta & -\sin \Theta & 0 \\ \sin \Theta & \cos \Theta & 0 \\ 0 & 0 & I \end{bmatrix} \begin{bmatrix} V^T & 0 \\ 0 & U_1^T \\ 0 & U_2^T \end{bmatrix},$$

which is the CS decomposition [13, 24] (see also [Nick Higham's blog post](#)) of $e^t \begin{bmatrix} 0 & -B^T \\ B & 0 \end{bmatrix} \in O(n)$.

In the CS decomposition of a matrix, the θ_i are the *principal angles* between the subspace spanned by the first k columns of the matrix and the coordinate subspace spanned by the first k standard basis vectors (i.e., the first k columns of the identity matrix). Moreover, the $\cos \theta_i$ are the singular values of the $k \times k$ Grammian $I_{n,k}^T Y$, where Y is the $n \times k$ matrix consisting of the first k columns.

More generally, if Y_1 and Y_2 are $n \times k$ matrices with orthonormal columns, the cosines of the principal angles $\theta_1, \dots, \theta_k$ between the subspaces they span are the singular values of the Grammian $Y_1^T Y_2$.

In other words,

Theorem 4.6.2. Suppose Y_1 and Y_2 are $n \times k$ matrices with orthonormal columns representing points $[Y_1], [Y_2] \in \text{Gr}(k, \mathbb{R}^n)$. Then

$$d([Y_1], [Y_2]) = \sqrt{\sum_{i=1}^k \theta_i^2},$$

where the $\cos \theta_i$ are the singular values of $Y_1^T Y_2$.

In fact, the singular value decomposition of $Y_1^T Y_2$ actually tells you what this geodesic is. Say

$$Y_1^T Y_2 = U(\cos \Theta) V^T$$

is the singular value decomposition. I can think of U and V as change of basis matrices on $\text{col}(Y_1)$ and $\text{col}(Y_2)$, respectively; that is, $[Y_1] = [Y_1 U]$ and $[Y_2] = [Y_2 V]$ and the columns of $Y_1 U$ and $Y_2 V$ give new orthonormal bases for the these respective subspaces (i.e., points on the Grassmannian). If I let u_i be the i th column of $Y_1 U$ and let v_j be the j th column of $Y_2 V$, then

$$\cos \Theta = U^T Y_1^T Y_2 V = (Y_1 U)^T (Y_2 V) = [\langle u_i, v_j \rangle]_{i,j}.$$

This tells us that the bases u_1, \dots, u_k for $[Y_1]$ and v_1, \dots, v_k for $[Y_2]$ are in some sense optimally aligned: in particular, they realize the principal angles between the subspaces, since the angle between u_i and v_i is θ_i . This is sketched in [Figure 4.11](#).

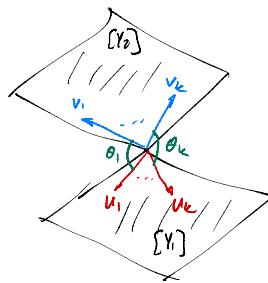


Figure 4.11: The aligned bases of two planes which realize the principal angles $\theta_1, \dots, \theta_k$ between the planes.

And now, the geodesic from $[Y_1]$ to $[Y_2]$ is simple:

Theorem 4.6.3. Let $u_i(t)$ be the rotation of each u_i to the corresponding v_i at constant speed so that the angle between $u_i = u_i(0)$ and $u_i(t)$ is $t\theta_i$. This ensures that each $v_i = u_i(1)$. If $\gamma(t)$ is the subspace spanned by $u_1(t), \dots, u_k(t)$, which can be represented by the $n \times k$ orthonormal matrix $[u_1(t) \ \dots \ u_k(t)]$, then γ is a minimizing geodesic in $\text{Gr}(k, \mathbb{R}^n)$ between $\gamma(0) = [Y_1]$ and $\gamma(1) = [Y_2]$.

For example, $\text{Gr}(1, \mathbb{R}^n) = \mathbb{RP}^{n-1}$ consists of lines through the origin, and this tells us that the geodesic between two lines is just the steady-speed rotation of one to the other.

Example 4.6.4. In general, I can get random points in $\text{Gr}(k, \mathbb{R}^n)$ by applying Gram–Schmidt to $n \times k$ matrices with standard Gaussian entries.

With $k = 26$ and $n = 57$ I generated Y_1 and Y_2 in this way; they’re visualized in the first row of [Figure 4.12](#) by having the value in each entry specify a color. The second row of [Figure 4.12](#) shows the aligned representatives Y_1U and Y_2V . Using [Theorem 4.6.2](#), $d([Y_1], [Y_2]) \approx 4.692$.

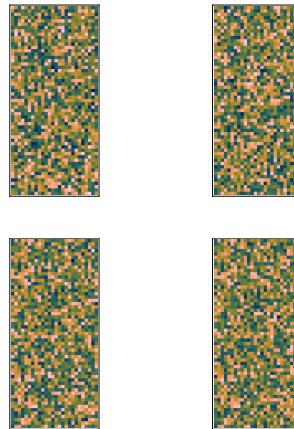


Figure 4.12: Two points in $\text{Gr}(k, \mathbb{R}^n)$ represented by matrices visualized in the first row, with the aligned representatives visualized in the second row.

In turn, [Figure 4.13](#) shows a few equally-spaced points along the minimizing geodesic given by [Theorem 4.6.3](#) connecting these two points in the Grassmannian.

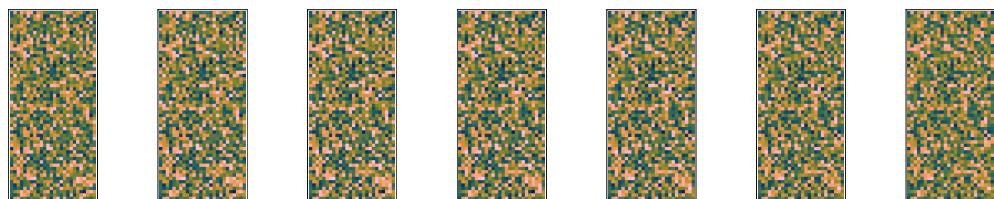


Figure 4.13: Seven equally-spaced points along the minimizing geodesic from $[Y_1]$ to $[Y_2]$ in $\text{Gr}(26, \mathbb{R}^{57})$.

Here is a video showing the full geodesic:

Video Visit the URL below to view a video:

<https://player.vimeo.com/video/1078821889>



Using [Theorem 4.6.2](#) it is straightforward to compute distances between points in the Grassmannian. [Figure 4.14](#) shows the histogram of distances between 10,000 pairs of random points in $\text{Gr}(26, \mathbb{R}^{57})$, which fits quite well to the Gaussian distribution with mean 4.844 and standard deviation 0.0788. Note that the maximum possible distance is $\sqrt{26} \frac{\pi}{2} \approx 8.0095$.

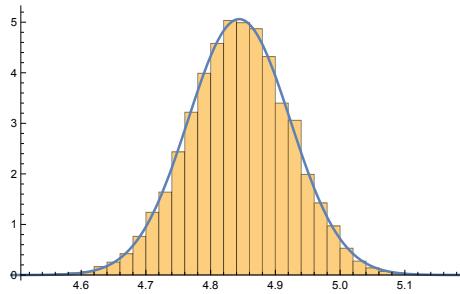


Figure 4.14: Histogram of distances between 10,000 pairs of random points on $\text{Gr}(26, \mathbb{R}^{57})$, together with the graph of the probability density function of the Gaussian distribution with mean 4.844 and standard deviation 0.0788.

4.7 The Curvature Tensor

Given the name, you might expect that some of the key ideas in Riemannian geometry are due to Bernhard Riemann. Indeed, he explained a way of thinking about what eventually came to be called Riemannian manifolds and curvature in his *Habilitationsschrift* from 1854 [25].

His idea for how to define curvature in a manifold was something like the following. Let $p \in M$ and consider $\sigma \subset T_p M$, a 2-dimensional subspace. Each line through the origin in σ exponentiates out to give a geodesic in M , and by taking all such geodesics we get a surface S in M containing p and tangent to σ at p . The Riemannian metric on M induces a Riemannian metric on S .

Gauss had showed in 1827 [11] that the Gauss curvature of a surface depends only on its First Fundamental Form (a.k.a., Riemannian metric); Riemann called this the curvature at p , denoted $K(p, \sigma)$. These days we call this curvature the *sectional curvature* at p with respect to σ .

That said, Riemann did not give any way to compute this curvature and it took quite a long time to get a rigorous, modern definition. Unfortunately, Riemann's intuition, which hopefully makes sense, is not really evident in the modern definition, which seems very abstract.

That said, I'm going to give you the modern definitions and try to connect it back to the geometric intuition as much as I can.

Definition 4.7.1. The (*Riemann*) *curvature tensor* R on a Riemannian manifold M is a map $R: \mathfrak{X}(M) \times \mathfrak{X}(M) \times \mathfrak{X}(M) \rightarrow \mathfrak{X}(M)$ given by

$$R(X, Y)Z = \nabla_X \nabla_Y Z - \nabla_Y \nabla_X Z - \nabla_{[X, Y]} Z, \quad (4.20)$$

where ∇ is the Levi-Civita connection associated with the Riemannian metric.

Remark 4.7.2. Some books, including do Carmo [7], define the curvature tensor to be

$$\nabla_Y \nabla_X Z - \nabla_X \nabla_Y Z + \nabla_{[X, Y]} Z;$$

notice that this is $-R(X, Y)Z$ as we've defined it.

Remark 4.7.3. We can see immediately that $R(Y, X)Z = -R(X, Y)Z$, so R is skew-symmetric in the first two factors.

Remark 4.7.4. Given $X, Y \in \mathfrak{X}(M)$, it is often convenient to think of $R(X, Y): \mathfrak{X}(M) \rightarrow \mathfrak{X}(M)$, sometimes called the *curvature transformation*. Note that

$$R(X, Y) = [\nabla_X, \nabla_Y] - \nabla_{[X, Y]}.$$

If we tried to define a similar curvature transformation for the Lie derivative we would have

$$[L_X, L_Y] - L_{[X, Y]} = 0$$

since

$$\begin{aligned} L_X L_Y Z - L_Y L_X Z - L_{[X, Y]} Z &= [X, [Y, Z]] - [Y, [X, Z]] - [[X, Y], Z] \\ &= -[[Y, Z], X] - [[Z, X], Y] - [[X, Y], Z] = 0 \end{aligned}$$

by the Jacobi identity [Proposition 1.6.4\(iii\)](#).

Remark 4.7.5. In local coordinates, $\left[\frac{\partial}{\partial x_i}, \frac{\partial}{\partial x_j} \right] = 0$, so

$$R \left(\frac{\partial}{\partial x_i}, \frac{\partial}{\partial x_j} \right) \frac{\partial}{\partial x_k} = \left(\nabla_{\partial/\partial x_i} \nabla_{\partial/\partial x_j} - \nabla_{\partial/\partial x_j} \nabla_{\partial/\partial x_i} \right) \frac{\partial}{\partial x_k},$$

so the curvature tensor is measuring the extent to which iterated covariant differentiation is non-commutative.

Example 4.7.6. In \mathbb{R}^n with the Euclidean metric, $R \equiv 0$. To see this, write $Z = (z_1, \dots, z_n)$ and recall from [Example 4.2.3](#) that covariant differentiation works componentwise:

$$\nabla_X Z = \sum_k X(z_k) \frac{\partial}{\partial x_k}. \quad (4.21)$$

Hence,

$$\nabla_X \nabla_Y Z = \sum_k X(Y(z_k)) \frac{\partial}{\partial x_k}$$

and similarly for $\nabla_Y \nabla_X Z$, so we get

$$\begin{aligned} R(X, Y)Z &= \nabla_X \nabla_Y Z - \nabla_Y \nabla_X Z - \nabla_{[X, Y]} Z \\ &= \sum_k [X(Y(z_k)) - Y(X(z_k)) - [X, Y](z_k)] \frac{\partial}{\partial x_k} \\ &= 0, \end{aligned}$$

so Euclidean space is reassuringly as un-curved as it is possible to be, and suggests that this notion of curvature is some sort of measure of how much the metric on M deviates from being Euclidean.

Example 4.7.7. Recall our vector fields X, Y, Z on S^3 defined in [Section 1.8](#) as

$$X(x) = xi, \quad Y(x) = xj, \quad Z(x) = xk$$

at each $x \in S^3$.⁵ We showed in [Section 1.8](#) that

$$[X, Y] = 2Z, \quad [Y, Z] = 2X, \quad [Z, X] = 2Y. \quad (4.22)$$

S^3 inherits its standard metric from the Euclidean metric on \mathbb{R}^4 , and the corresponding Levi-Civita connection ∇ on S^3 is just the restriction to S^3 (via orthogonal projection) of the Euclidean connection on \mathbb{R}^4 , which I'll denote as $\nabla^{\mathbb{R}^4}$. In other words, for $U, V \in \mathfrak{X}(S^3)$, $(\nabla_U V)(p)$ is the orthogonal projection of $(\nabla_U^{\mathbb{R}^4} V)(p)$ to $T_x S^3 = x^\perp$.

If I write out X in Euclidean coordinates, I get

$$X(x) = xi = (x_1 + x_2 i + x_3 j + x_4 k)i = -x_2 + x_1 i + x_4 j - x_3 k = -x_2 \frac{\partial}{\partial x_1} + x_1 \frac{\partial}{\partial x_2} + x_4 \frac{\partial}{\partial x_3} - x_3 \frac{\partial}{\partial x_4},$$

and similarly

$$\begin{aligned} Y(x) &= -x_3 \frac{\partial}{\partial x_1} - x_4 \frac{\partial}{\partial x_2} + x_1 \frac{\partial}{\partial x_3} + x_2 \frac{\partial}{\partial x_4} \\ Z(x) &= -x_4 \frac{\partial}{\partial x_1} + x_3 \frac{\partial}{\partial x_2} - x_2 \frac{\partial}{\partial x_3} + x_1 \frac{\partial}{\partial x_4}. \end{aligned}$$

So then, again using (4.21), we have that

$$\begin{aligned} \nabla_X^{\mathbb{R}^4} Y &= \nabla_X^{\mathbb{R}^4} \left(-x_3 \frac{\partial}{\partial x_1} - x_4 \frac{\partial}{\partial x_2} + x_1 \frac{\partial}{\partial x_3} + x_2 \frac{\partial}{\partial x_4} \right) \\ &= -X(x_3) \frac{\partial}{\partial x_1} - X(x_4) \frac{\partial}{\partial x_2} + X(x_1) \frac{\partial}{\partial x_3} + X(x_2) \frac{\partial}{\partial x_4} \\ &= -x_4 \frac{\partial}{\partial x_1} + x_3 \frac{\partial}{\partial x_2} - x_2 \frac{\partial}{\partial x_3} + x_1 \frac{\partial}{\partial x_4} \\ &= Z, \end{aligned}$$

⁵I'm switching from p to x as the name of the point in S^3 because I'm going to think of points in S^3 as points in \mathbb{R}^4 and write them as $x = (x_1, x_2, x_3, x_4)$, where the x_i are coordinate functions. This will make it more natural to write things like $\frac{\partial}{\partial x_i}(x_i) = 1$ rather than calling the coordinate functions p_i and trying to remember that $\frac{\partial}{\partial x_i}(p_i) = 1$.

which is already tangent to S^3 , so orthogonal projection just gives Z again, and we conclude that $\nabla_X Y = Z$.

By similar reasoning, we can compute

$$\begin{array}{lll} \nabla_X X = 0 & \nabla_X Y = Z & \nabla_X Z = -Y \\ \nabla_Y X = -Z & \nabla_Y Y = 0 & \nabla_Y Z = X \\ \nabla_Z X = Y & \nabla_Z Y = -X & \nabla_Z Z = 0 \end{array} \quad (4.23)$$

Therefore, we can use (4.22) and (4.23) to see that

$$\begin{aligned} R(X, Y)X &= \nabla_X \nabla_Y X - \nabla_Y \nabla_X X - \nabla_{[X, Y]} X \\ &= -\nabla_X Z - \nabla_Y 0 - \nabla_{2Z} X \\ &= Y - 2Y \\ &= -Y, \end{aligned}$$

that

$$\begin{aligned} R(X, Y)Y &= \nabla_X \nabla_Y Y - \nabla_Y \nabla_X Y - \nabla_{[X, Y]} Y \\ &= \nabla_X 0 - \nabla_Y Z - \nabla_{2Z} Y \\ &= X + 2X \\ &= X, \end{aligned} \quad (4.24)$$

and that

$$\begin{aligned} R(X, Y)Z &= \nabla_X \nabla_Y Z - \nabla_Y \nabla_X Z - \nabla_{[X, Y]} Z \\ &= \nabla_X X + \nabla_Y Y - \nabla_{2Z} Z \\ &= 0 + 0 - 0 \\ &= 0. \end{aligned}$$

Therefore, for each $x \in S^3$ the transformation $R(X, Y) : T_x S^3 \rightarrow T_x S^3$ can be interpreted as the composition of orthogonal projection to the 2-plane spanned by X and Y and a clockwise rotation of this plane by 90° .

Exercise 4.7.8. Work out similar interpretations of $R(X, Z)$ and $R(Y, Z)$ on S^3 .

Despite having called it the *curvature tensor*, I have not yet showed that R is a tensor field. To be a proper tensor field, the value of $R(X, Y)Z$ at a point would need to depend *only* on the values of X, Y, Z at the point; given all the derivatives involved, you might expect that it would depend on the values of X, Y, Z in a neighborhood of the point.

Proposition 4.7.9. *The curvature tensor $R(X, Y)Z$ on a Riemannian manifold M is multilinear in X, Y, Z over the algebra $C^\infty(M)$ of smooth functions on M .*

Of course, (4.20) shows that $R(X, Y)Z$ is multilinear over \mathbb{R} .

Exercise 4.7.10. Show that multilinearity over $C^\infty(M)$ implies that the value of $R(X, Y)Z$ at $p \in M$ depends only on the values of X, Y, Z at p , and not on their values in any neighborhood of p .

Proof of Proposition 4.7.9. Suppose $f \in C^\infty(M)$. Then, repeatedly using the parts of Definition 4.2.2,

$$\begin{aligned} R(fX, Y)Z &= \nabla_{fX}\nabla_Y Z - \nabla_Y\nabla_{fX}Z - \nabla_{[fX, Y]}Z \\ &= f\nabla_X\nabla_Y Z - \nabla_Y(f\nabla_X Z) - \nabla_{f[X, Y] - Y(f)X}Z \\ &= f\nabla_X\nabla_Y Z - Y(f)\nabla_X Z - f\nabla_Y\nabla_X Z - f\nabla_{[X, Y]}Z + Y(f)\nabla_X Z \\ &= f(\nabla_X\nabla_Y Z - \nabla_Y\nabla_X Z - \nabla_{[X, Y]}Z) \\ &= fR(X, Y)Z. \end{aligned}$$

Hence, $R(X, Y)Z$ is multilinear in X over $C^\infty(M)$. Since $R(X, Y)Z = -R(Y, X)Z$, this shows that it is also multilinear in Y over $C^\infty(M)$.

Finally,

$$\begin{aligned} R(X, Y)(fZ) &= \nabla_X\nabla_Y fZ - \nabla_Y\nabla_X fZ - \nabla_{[X, Y]}fZ \\ &= \nabla_X(Y(f)Z + f\nabla_Y Z) - \nabla_Y(X(f)Z + f\nabla_X Z) - [X, Y](f)Z - f\nabla_{[X, Y]}Z \\ &= X(Y(f))Z + Y(f)\nabla_X Z + X(f)\nabla_Y Z + f\nabla_X\nabla_Y Z \\ &\quad - Y(X(f))Z - X(f)\nabla_Y Z - Y(f)\nabla_X Z - f\nabla_Y\nabla_X Z \\ &\quad - X(Y(f))Z + Y(X(f))Z - f\nabla_{[X, Y]}Z \\ &= f(\nabla_X\nabla_Y Z - \nabla_Y\nabla_X Z - \nabla_{[X, Y]}Z) \\ &= fR(X, Y)Z, \end{aligned}$$

showing the linearity of $R(X, Y)Z$ in Z over $C^\infty(M)$ and completing the proof. \square

Proposition 4.7.11 (Bianchi Identity). *For M a Riemannian manifold and $X, Y, Z \in \mathfrak{X}(M)$,*

$$R(X, Y)Z + R(Y, Z)X + R(Z, X)Y = 0.$$

Proof. Since ∇ is symmetric, we know that $\nabla_A B - \nabla_B A = [A, B]$. Therefore,

$$\nabla_Y Z = [Y, Z] + \nabla_Z Y$$

and

$$\nabla_{[X, Y]}Z = [[X, Y], Z] + \nabla_Z[X, Y].$$

Therefore,

$$\begin{aligned} R(X, Y)Z &= \nabla_X\nabla_Y Z - \nabla_Y\nabla_X Z - \nabla_{[X, Y]}Z \\ &= \nabla_X([Y, Z] + \nabla_Z Y) - \nabla_Y\nabla_X Z - [[X, Y], Z] - \nabla_Z[X, Y], \end{aligned} \tag{4.25}$$

and similarly

$$R(Y, Z)X = \nabla_Y([Z, X] + \nabla_X Z) - \nabla_Z\nabla_Y X - [[Y, Z], X] - \nabla_X[Y, Z] \tag{4.26}$$

$$R(Z, X)Y = \nabla_Z([X, Y] + \nabla_Y X) - \nabla_X\nabla_Z Y - [[Z, X], Y] - \nabla_Y[Z, X]. \tag{4.27}$$

If we add (4.25), (4.26), and (4.27), all the connection terms cancel out and we're left with

$$R(X, Y)Z + R(Y, Z)X + R(Z, X)Y = -([[X, Y], Z] + [[Y, Z], X] + [[Z, X], Y]) = 0$$

by the Jacobi identity Proposition 1.6.4(iii). \square

We've defined R as a map $\mathfrak{X}(M) \times \mathfrak{X}(M) \times \mathfrak{X}(M) \rightarrow \mathfrak{X}(M)$; that is, $R(X, Y)Z \in \mathfrak{X}(M)$. We can interpret this as a $(1, 3)$ -tensor field as follows: at each $p \in M$ we get a map $(T_p M)^* \times T_p M \times T_p M \times T_p M \rightarrow \mathbb{R}$ given by

$$(\tau, X, Y, Z) = \tau(R(X, Y)Z).$$

(Note that this being well-defined depends essentially on [Exercise 4.7.10](#).)

Of course, we've defined everything in the presence of a Riemannian metric g , and we can always convert vectors into linear functionals with inner products: $T \leftrightarrow g(\cdot, T)$ for any $T \in \mathfrak{X}(M)$, and so we can convert R into a $(0, 4)$ -tensor field—that is, a $C^\infty(M)$ -multilinear map $\mathfrak{X}(M) \times \mathfrak{X}(M) \times \mathfrak{X}(M) \times \mathfrak{X}(M) \rightarrow \mathbb{R}$ by mapping

$$(X, Y, Z, T) \mapsto g(R(X, Y)Z, T).$$

Proposition 4.7.12 (Symmetries of the Curvature Tensor). *For $X, Y, Z, T \in \mathfrak{X}(M)$,*

- (i) (*Bianchi Identity*) $g(R(X, Y)Z, T) + g(R(Y, Z)X, T) + g(R(Z, X)Y, T) = 0$.
- (ii) (*Skew-Symmetry*) $g(R(X, Y)Z, T) = -g(R(Y, X)Z, T)$
- (iii) (*Skew-Symmetry*) $g(R(X, Y)Z, T) = -g(R(X, Y)T, Z)$
- (iv) (*Interchange Symmetry*) $g(R(X, Y)Z, T) = g(R(Z, T)X, Y)$

Proof. (i) follows immediately from [Proposition 4.7.11](#).

(ii) follows from the fact that $R(X, Y) = -R(Y, X)$, which we saw was a direct consequence of the definition.

For (iii), observe that

$$(XY - YX - [X, Y])(g(Z, T)) = 0$$

since $XY - YX = [X, Y]$. Now, if we repeatedly use [Corollary 4.3.5](#) we see that

$$\begin{aligned} XY(g(Z, T)) &= X(g(\nabla_Y Z, T) + g(Z, \nabla_Y T)) = g(\nabla_X \nabla_Y Z, T) + g(\nabla_Y Z, \nabla_X T) + g(\nabla_X Z, \nabla_Y T) + g(Z, \nabla_X \nabla_Y T) \\ YX(g(Z, T)) &= Y(g(\nabla_X Z, T) + g(Z, \nabla_X T)) = g(\nabla_Y \nabla_X Z, T) + g(\nabla_X Z, \nabla_Y T) + g(\nabla_Y Z, \nabla_X T) + g(Z, \nabla_Y \nabla_X T) \\ [X, Y](g(Z, T)) &= g(\nabla_{[X, Y]} Z, T) + g(Z, \nabla_{[X, Y]} T). \end{aligned}$$

Subtracting the second and third lines from the first yields

$$\begin{aligned} 0 &= (XY - YX - [X, Y])(g(Z, T)) \\ &= g(\nabla_X \nabla_Y Z - \nabla_Y \nabla_X Z - \nabla_{[X, Y]} Z, T) + g(Z, \nabla_X \nabla_Y T - \nabla_Y \nabla_X T - \nabla_{[X, Y]} T) \\ &= g(R(X, Y)Z, T) + g(R(X, Y)T, Z), \end{aligned}$$

which implies (iii).

For (iv), note that the Bianchi identity implies that

$$\begin{aligned} g(R(X, Y)Z, T) + g(R(Y, Z)X, T) + g(R(Z, X)Y, T) &= 0 \\ g(R(Y, Z)T, X) + g(R(Z, T)Y, X) + g(R(T, Y)Z, X) &= 0 \\ g(R(Z, T)X, Y) + g(R(T, X)Z, Y) + g(R(X, Z)T, Y) &= 0 \\ g(R(T, X)Y, Z) + g(R(X, Y)T, Z) + g(R(Y, T)X, Z) &= 0. \end{aligned}$$

Adding the first two lines and subtracting the third and fourth and using (ii) and (iii) to cancel four pairs of terms and simplify the remaining terms yields

$$2g(R(X, Y)Z, T) - 2g(R(Z, T)X, Y) = 0,$$

which implies the result. \square

In local coordinates, if we write $X_i = \frac{\partial}{\partial x_i}$, then there exist smooth functions R_{ijk}^s so that

$$R(X_j, X_k)X_i = \sum_s R_{ijk}^s X_s.$$

The functions R_{ijk}^s are usually called the components of R in the given local coordinates. If we have vector fields U, V, W with local coordinate expressions

$$U = \sum_j u^j X_j, \quad V = \sum_k v^k X_k, \quad W = \sum_i w^i X_i$$

(note that now I am being more careful about whether indices are superscripted or subscripted), then

$$R(U, V)W = \sum_{i,j,k,s} R_{ijk}^s u^j v^k w^i X_s.$$

(In Einstein notation, this gets written as $R(U, V)W = R_{ijk}^s u^j v^k w^i X_s$, where summation always happens over repeated indices, and repeated indices should appear in up/down pairs.)

Exercise 4.7.13. Show that

$$R_{ijk}^s = \frac{\partial \Gamma_{ik}^s}{\partial x_j} - \frac{\partial \Gamma_{jk}^s}{\partial x_i} + \sum_m (\Gamma_{ik}^m \Gamma_{jm}^s - \Gamma_{jk}^m \Gamma_{im}^s).$$

When we lower an index and think of R as a $(0, 4)$ -tensor, then we get components

$$R_{ijks} := g(R(X_k, X_s)X_i, X_j) = \left\langle \sum_m R_{iks}^m X_m, X_j \right\rangle = \sum_m R_{iks}^m g_{mj}.$$

Then Proposition 4.7.12 implies the following symmetries for the R_{ijks} :

Corollary 4.7.14. (i) $R_{ijks} + R_{iksj} + R_{isjk} = 0$

(ii) $R_{ijks} = -R_{jiks}$

(iii) $R_{ijks} = -R_{ijsk}$

(iv) $R_{ijks} = R_{ksij}$.

In particular, recall that we proved the interchange symmetry (iv) from the other three, and the first three turn out to be a complete list of symmetries of the curvature tensor, so R is completely specified in local coordinates by the choice of $\frac{n^2(n-1)^2}{12}$ independent functions R_{ijks} .

4.8 Sectional Curvature

We are now in a position to define sectional curvature, which will agree with Riemann's original notion of curvature. To do so, we introduce one piece of notation: for V an inner product space and $u, v \in V$, let $\|u \wedge v\|$ be the area of the parallelogram spanned by u and v ; that is,

$$\|u \wedge v\|^2 = \|u\|^2 \|v\|^2 - \langle u, v \rangle^2. \quad (4.28)$$

Remark 4.8.1. The reason for the notation is that $\|u \wedge v\|$ agrees with the norm on $\wedge^2(V)$ induced by the inner product defined in HW 2 Problem 4.

Remark 4.8.2. When (M, g) is a Riemannian manifold and $V = T_p M$, the corresponding inner product is g_p , so that

$$\|u \wedge v\|^2 = \|u\|^2 \|v\|^2 - g_p(u, v)^2 = g_p(u, u)g_p(v, v) - g_p(u, v)^2.$$

Definition 4.8.3. Let (M, g) be a Riemannian manifold, let $p \in M$ and let $\sigma \subset T_p M$ be a 2-dimensional subspace, and let $\{u, v\}$ be any basis for σ . Then the number

$$K(\sigma) := \frac{g_p(R(u, v)v, u)}{\|u \wedge v\|^2}$$

is the *sectional curvature* of σ at p .

Since the right hand side involves a choice of basis for σ , it is not at all clear yet that $K(\sigma)$ is well-defined.

Proposition 4.8.4. *With the same setup as Definition 4.8.3, $K(\sigma)$ does not depend on the choice of $u, v \in \sigma$.*

Proof. Since we haven't shown independence from the choice of u and v yet, write

$$K(u, v) = \frac{g_p(R(u, v)v, u)}{\|u \wedge v\|^2}.$$

Then $K(u, v)$ has the following symmetries:

- (i) $K(u, v) = K(v, u)$ by skew-symmetry of R (Proposition 4.7.12(ii) and (iii)) and inspection of (4.28).
- (ii) $K(u, v) = K(\lambda u, v)$ for any $\lambda \in \mathbb{R}$ by multilinearity of R and homogeneity of $\|u \wedge v\|$.
- (iii) $K(u, v) = K(u + \lambda v, v)$ by the multilinearity and skew-symmetry of R and \wedge .

The transformations $(u, v) \mapsto (v, u)$, $(u, v) \mapsto (\lambda u, v)$, and $(u, v) \mapsto (u + \lambda v, v)$ correspond to the 2×2 matrices $\begin{bmatrix} 0 & 1 \\ 1 & 0 \end{bmatrix}$, $\begin{bmatrix} \lambda & 0 \\ 0 & 1 \end{bmatrix}$, and $\begin{bmatrix} 1 & \lambda \\ 0 & 1 \end{bmatrix}$, respectively. These matrices generate $GL_2(\mathbb{R})$, so we've just shown that K is invariant under the $GL_2(\mathbb{R})$ action on σ . Of course, $GL_2(\mathbb{R})$ transforms any basis into any other basis, so this shows that K is basis-independent. \square

Example 4.8.5. Continuing Example 4.7.7, let $X, Y, Z \in \mathfrak{X}(S^3)$ be our familiar vector fields with $X(x) = xi$, $Y(x) = xj$, $Z(x) = xk$. At a point $x \in S^3$, let $\sigma \subset T_x S^3$ be the plane spanned by X and Y . The parallelogram spanned by X and Y is a unit square, so $\|X \wedge Y\|^2 = 1$ and hence

$$K(\sigma) = \frac{g_x(R(X, Y)Y, X)}{\|X \wedge Y\|^2} = g_x(X, X) = 1$$

by (4.24) and the fact that X is a unit vector.

More generally, let $\sigma \subset T_x S^3$ be any 2-dimensional subspace and let $U = aX + bY + cZ$ and $V = rX + sY + tZ$ be an orthonormal basis for σ . Then we can use (4.24) and analogous expressions from Example 4.7.7 and Exercise 4.7.8 to conclude that

$$g(R(U, V)V, U) = a^2s^2 + a^2t^2 + b^2r^2 + b^2t^2 + c^2r^2 + c^2s^2 - 2abrs - 2acrt - 2bcst.$$

Since U and V are perpendicular, $0 = g(U, V) = ar + bs + ct$, so we conclude that

$$abrs + acrt = ar(bs + ct) = ar(-ar) = -a^2r^2$$

$$abrs + bcst = bs(ar + ct) = bs(-bs) = -b^2s^2$$

$$acrt + bcst = ct(ar + bs) = ct(-ct) = -c^2t^2,$$

so

$$2abrs + 2acrt + 2bcst = -a^2r^2 - b^2s^2 - c^2t^2$$

and therefore

$$g(R(U, V)V, U) = a^2(r^2 + s^2 + t^2) + b^2(r^2 + s^2 + t^2) + c^2(r^2 + s^2 + t^2) = 1$$

since $1 = \|U\|^2 = a^2 + b^2 + c^2$ and $1 = \|V\|^2 = r^2 + s^2 + t^2$. Moreover, the parallelogram spanned by U and V is again a unit square, so $\|U \wedge V\|^2 = 1$ and we conclude that

$$K(\sigma) = g(R(U, V)V, U) = 1.$$

Since σ and x were arbitrary, we see that S^3 has constant sectional curvature.

Example 4.8.6. Let's work out the sectional curvature for $\text{Aff}^+(\mathbb{R}) \cong H$, building on Example 4.3.10. In this case, $T_{(x,y)}H$ is already 2-dimensional, so the only 2-dimensional subspace is $T_{(x,y)}H$. Therefore, in this case (and, more generally, on any surface), we can interpret the sectional curvature as a function $K(x, y) = K(T_{(x,y)}H)$ on H , which will be the same thing as Gauss curvature. Of course, $X = \frac{\partial}{\partial x}$ and $Y = \frac{\partial}{\partial y}$ serve as a basis for $T_{(x,y)}H$ at each point. Moreover,

$$\|X \wedge Y\|^2 = g_{(x,y)}(X, X)g_{(x,y)}(Y, Y) - g_{(x,y)}(X, Y)g_{(x,y)}(Y, X) = \frac{1}{y^2} \frac{1}{y^2} - 0 = \frac{1}{y^4},$$

so

$$K(x, y) = \frac{g(R(X, Y)Y, X)}{\|X \wedge Y\|^2} = y^4 g(R(X, Y)Y, X). \quad (4.29)$$

By definition,

$$\begin{aligned} R(X, Y)Y &= \nabla_X \nabla_Y Y - \nabla_Y \nabla_X Y - \nabla_{[X, Y]} Y \\ &= -\nabla_X \frac{1}{y} Y + \nabla_Y \frac{1}{y} X - \nabla_0 Y \\ &= -X \left(\frac{1}{y} \right) Y - \frac{1}{y} \nabla_X Y + Y \left(\frac{1}{y} \right) X + \frac{1}{y} \nabla_Y X - 0 \\ &= 0 + \frac{1}{y^2} X - \frac{1}{y^2} X - \frac{1}{y^2} X \\ &= -\frac{1}{y^2} X. \end{aligned}$$

by repeatedly using (4.13) and recalling that $[X, Y] = 0$. Combining this with (4.29) we see that

$$K(x, y) = y^4 g(R(X, Y)Y, X) = -y^2 g(X, X) = -y^2 \frac{1}{y^2} = -1$$

by Example 4.1.7. That is, $\text{Aff}^+(\mathbb{R}) \cong H$ has constant sectional/Gauss curvature -1 . Indeed, this is also called the Poincaré upper half-plane model of the hyperbolic plane, as stated in Example 2.3.22 but now much more fully justified.

4.8.1 Sectional Curvature of Bi-Invariant Metrics

The last two examples have computed sectional curvatures of left-invariant metrics on Lie groups. In the case of S^3 (where the metric was actually bi-invariant), we got constant positive sectional curvatures; in the case of $\text{Aff}^+(\mathbb{R}) \cong H$, we got constant negative sectional curvature. In general Lie groups won't have constant sectional curvature, but it turns out that bi-invariant metrics will always have non-negative sectional curvature. So the phenomenon of negative sectional curvature that we saw with $\text{Aff}^+(\mathbb{R})$ was only possible because this was not a bi-invariant metric (as we saw in Example 4.5.6).

Indeed, sectional curvatures for bi-invariant metrics take a particularly simple form:

Theorem 4.8.7. *Let G be a Lie group with a bi-invariant metric. Let X and Y be orthonormal left-invariant vector fields on G . Then the sectional curvature $K(\sigma)$ of G with respect to the plane σ spanned by X and Y is given by*

$$K(\sigma) = \frac{1}{4} \| [X, Y] \|^2.$$

In other words, the sectional curvature of a Lie group with a bi-invariant metric satisfies $K(\sigma) \geq 0$, with equality if and only if σ is spanned by vectors X and Y that commute in the sense that $[X, Y] = 0$.

Note that the restriction that X and Y be orthonormal and left-invariant is not restrictive: if $\sigma \subset T_h G$, then we can just take some orthonormal basis for σ and then push it around all of G by the differential of left-translation to get a pair of left-invariant vector fields; the left-invariance of the metric will ensure the vector fields are orthonormal everywhere.

Proof. As we saw in the proof of Theorem 4.5.11, $\nabla_V V = 0$ for any left-invariant vector field V . So if X and Y are left-invariant we have

$$0 = \nabla_{X+Y}(X + Y) = \nabla_X X + \nabla_X Y + \nabla_Y X + \nabla_Y Y = \nabla_X Y + \nabla_Y X.$$

On the other hand, since ∇ is symmetric we know that

$$\nabla_X Y - \nabla_Y X = [X, Y].$$

Adding the two previous equations yields

$$\nabla_X Y = \frac{1}{2} [X, Y] \tag{4.30}$$

for any left-invariant vector fields X, Y on a Lie group with a bi-invariant metric.

Now, if X, Y, Z are left-invariant vector fields on G , then we can repeatedly apply (4.30) to see that

$$R(X, Y)Z = \nabla_X \nabla_Y Z - \nabla_Y \nabla_X Z - \nabla_{[X, Y]} Z = \frac{1}{4}[X, [Y, Z]] - \frac{1}{4}[Y, [X, Z]] - \frac{1}{2}[[X, Y], Z].$$

After rearranging a bit, this implies

$$4R(X, Y)Z = [X, [Y, Z]] + [Y, [Z, X]] + 2[Z, [X, Y]] \quad (4.31)$$

On the other hand, the Jacobi identity (Proposition 1.6.4(iii)) implies

$$0 = -[X, [Y, Z]] - [Y, [Z, X]] - [Z, [X, Y]] \quad (4.32)$$

Adding (4.31) and (4.32) and dividing by 4 yields

$$R(X, Y)Z = \frac{1}{4}[Z, [X, Y]]$$

for any left-invariant vector fields X, Y, Z .

Therefore, the full curvature tensor takes the form

$$g(R(X, Y)Z, T) = \frac{1}{4}g([Z, [X, Y]], T) = \frac{1}{4}g([X, Y], [T, Z])$$

for any left-invariant vector fields X, Y, Z, T by Proposition 4.5.3.

Finally, then, if X and Y are orthonormal left-invariant vector fields spanning σ , then $\|X \wedge Y\|^2 = 1$ and hence

$$K(\sigma) = g(R(X, Y)Y, X) = \frac{1}{4}g([X, Y], [X, Y]) = \frac{1}{4}\|[X, Y]\|^2,$$

as claimed. \square

Example 4.8.8. On S^3 , we know that

$$[X, Y] = 2Z, \quad [Y, Z] = 2X, \quad [Z, X] = 2Y,$$

so Theorem 4.8.7 implies that, e.g., the sectional curvature of the plane spanned by X and Y is

$$K(\sigma) = \frac{1}{4}\|[X, Y]\|^2 = \frac{1}{4}\|2Z\|^2 = 1,$$

agreeing with our calculation in Example 4.8.5.

One way of interpreting Theorem 4.8.7 is that it says that sectional curvature of bi-invariant metrics on Lie groups is entirely calculable in the Lie algebra and hence is essentially algebraic. Since $\mathrm{SO}(3)$ and $\mathrm{SU}(2)$ have the same Lie algebra as S^3 (Proposition 3.4.3), they must also have constant sectional curvature $+1$ everywhere.

Example 4.8.9. Let's work out the sectional curvature of the bi-invariant metric on $\mathrm{SO}(4)$. If we take the left-invariant metric induced by $\frac{1}{2}$ times the Frobenius inner product on $T_I \mathrm{SO}(4)$, then this will be bi-invariant.⁶ Indeed, for $X, Y, Z \in \mathfrak{so}(4)$,

$$\begin{aligned} \frac{1}{2}\langle[X, Y], Z\rangle_{\mathrm{Fr}} &= \frac{1}{2}\mathrm{tr}([X, Y]^T Z) = -\frac{1}{2}\mathrm{tr}([X, Y]Z) = -\frac{1}{2}\mathrm{tr}(XYZ - YXZ) \\ &= -\frac{1}{2}\mathrm{tr}(YZX - YXZ) = -\frac{1}{2}\mathrm{tr}(Y[Z, X]) = \frac{1}{2}\mathrm{tr}(Y^T[Z, X]) = \frac{1}{2}\langle Y, [Z, X]\rangle_{\mathrm{Fr}} \end{aligned}$$

⁶It is traditional to scale by the factor of $\frac{1}{2}$ for reasons that will shortly become apparent.

$[\cdot, \cdot]$	H_1	H_2	E	F	X	Y
H_1	0	0	F	$-E$	Y	$-X$
H_2	0	0	$-F$	E	Y	$-X$
E	$-F$	F	0	$2(H_1 - H_2)$	0	0
F	E	$-E$	$-2(H_1 - H_2)$	0	0	0
X	$-Y$	$-Y$	0	0	0	$2(H_1 + H_2)$
Y	X	X	0	0	$-2(H_1 + H_2)$	0

Table 4.1: Lie brackets of the given basis of $\mathfrak{so}(4)$.

using the cyclic-invariance of trace and the fact that $[X, Y]$ and Y are skew-symmetric. Therefore, [Theorem 4.5.5](#) implies that the induced left-invariant metric is bi-invariant.

I claim that the following gives a basis for $\mathfrak{so}(4)$:

$$H_1 = \begin{bmatrix} 0 & 1 & 0 & 0 \\ -1 & 0 & 0 & 0 \\ 0 & 0 & 0 & 0 \\ 0 & 0 & 0 & 0 \end{bmatrix} \quad E = \begin{bmatrix} 0 & 0 & 1 & 0 \\ 0 & 0 & 0 & 1 \\ -1 & 0 & 0 & 0 \\ 0 & -1 & 0 & 0 \end{bmatrix} \quad X = \begin{bmatrix} 0 & 0 & 0 & 1 \\ 0 & 0 & 1 & 0 \\ 0 & -1 & 0 & 0 \\ -1 & 0 & 0 & 0 \end{bmatrix}$$

$$H_2 = \begin{bmatrix} 0 & 0 & 0 & 0 \\ 0 & 0 & 0 & 0 \\ 0 & 0 & 0 & 1 \\ 0 & 0 & -1 & 0 \end{bmatrix} \quad F = \begin{bmatrix} 0 & 0 & 0 & 1 \\ 0 & 0 & -1 & 0 \\ 0 & 1 & 0 & 0 \\ -1 & 0 & 0 & 0 \end{bmatrix} \quad Y = \begin{bmatrix} 0 & 0 & 1 & 0 \\ 0 & 0 & 0 & -1 \\ -1 & 0 & 0 & 0 \\ 0 & 1 & 0 & 0 \end{bmatrix}$$

Now,

$$[E, F] = EF - FE = \begin{bmatrix} 0 & 0 & 1 & 0 \\ 0 & 0 & 0 & 1 \\ -1 & 0 & 0 & 0 \\ 0 & -1 & 0 & 0 \end{bmatrix} \begin{bmatrix} 0 & 0 & 0 & 1 \\ 0 & 0 & -1 & 0 \\ 0 & 1 & 0 & 0 \\ -1 & 0 & 0 & 0 \end{bmatrix} - \begin{bmatrix} 0 & 0 & 0 & 1 \\ 0 & 0 & -1 & 0 \\ 0 & 1 & 0 & 0 \\ -1 & 0 & 0 & 0 \end{bmatrix} \begin{bmatrix} 0 & 0 & 1 & 0 \\ 0 & 0 & 0 & 1 \\ -1 & 0 & 0 & 0 \\ 0 & -1 & 0 & 0 \end{bmatrix}$$

$$= \begin{bmatrix} 0 & 1 & 0 & 0 \\ -1 & 0 & 0 & 0 \\ 0 & 0 & -1 & 0 \\ 0 & 0 & 1 & 0 \end{bmatrix} - \begin{bmatrix} 0 & -1 & 0 & 0 \\ 1 & 0 & 0 & 0 \\ 0 & 0 & 0 & 1 \\ 0 & 0 & -1 & 0 \end{bmatrix} = 2(H_1 - H_2)$$

and similarly we can compute all the other brackets, yielding the multiplication table given in [Table 4.1](#). Notice that

$$\|H_1\|^2 = \frac{1}{2}\langle H_1, H_1 \rangle_{\text{Fr}} = 1$$

and similarly for the other basis elements, so we see that $H_1, H_2, \frac{1}{\sqrt{2}}E, \frac{1}{\sqrt{2}}F, \frac{1}{\sqrt{2}}X, \frac{1}{\sqrt{2}}Y$ gives an orthonormal basis for $\mathfrak{so}(4)$.⁷

Hence, [Theorem 4.8.7](#) tells us that the plane $\sigma_{H_1 E}$ spanned by H_1 and E has sectional curvature

$$K(\sigma_{H_1 E}) = \frac{1}{4} \left\| \left[H_1, \frac{1}{\sqrt{2}}E \right] \right\|^2 = \frac{1}{4} \left\| \frac{1}{\sqrt{2}}F \right\|^2 = \frac{1}{4}$$

⁷This is why we scaled by $\frac{1}{2}$: we want H_1 and H_2 to be unit vectors.

$K(\sigma_{\cdot,\cdot})$	H_1	H_2	E	F	X	Y
H_1		0	1/4	1/4	1/4	1/4
H_2	0		1/4	1/4	1/4	1/4
E	1/4	1/4		1/2	0	0
F	1/4	1/4	1/2		0	0
X	1/4	1/4	0	0		1/2
Y	1/4	1/4	0	0	1/2	

Table 4.2: Sectional curvatures in $\text{SO}(4)$.

and in general we get the sectional curvatures given in [Table 4.2](#). Note that the diagonal entries would make no sense: the same vector twice spans a line, not a plane.

4.9 Ricci and Scalar Curvatures

Sectional curvature is an extremely powerful tool for describing the geometry of a manifold, and there is a huge amount of work out there about the impact it has on dynamics, the constraints it places on topology, and many other interactions it has with other concepts of interest. Just to give one example, the *sphere theorem* says that any complete, simply-connected n -manifold with sectional curvatures in the interval $[1, 4]$ is homeomorphic to the sphere S^n . This was conjectured by Hopf in 1932 [\[17\]](#) and proved independently in 1961 by Berger [\[4\]](#) and Klingenberg [\[19\]](#). In 2007, Brendle and Schoen [\[5\]](#) took this further and proved the *differentiable sphere theorem*, which says that any Riemannian manifold with these sectional curvature bounds is diffeomorphic to S^n with its standard smooth structure. These bounds are optimal: the complex projective space \mathbb{CP}^n with the Fubini–Study metric has sectional curvatures in the closed interval $[1, 4]$, and is not homeomorphic to a sphere unless $n = 1$.

That said, while it's conceptually somewhat simpler than the full Riemann curvature tensor, sectional curvature is still rather complicated. One way to simplify it is to take averages in the following sense: suppose $p \in M$ and $X \in T_p M$. Then we could take the average of all sectional curvatures of 2-planes σ containing X . Of course, this really only depends on the line spanned by X , so this assigns a number to tangent lines at points. Indeed, this is the basic idea of *Ricci curvature*.

Example 4.9.1. Let M be a Riemannian 3-manifold, let $p \in M$, and let $Z_1, Z_2, X \in T_p M$ be an orthonormal basis. The idea of Ricci curvature at X is to take the average of sectional curvatures of all 2-planes containing X . One way to do this is to compute

$$\frac{1}{2}K(X, Z_1) + \frac{1}{2}K(X, Z_2),$$

where I'm writing $K(X, Z_i)$ for the sectional curvature of the plane spanned by X and Z_i .

Of course, this really only computes the average over 2 specific planes, and seemingly depends on the choice of orthonormal basis Z_1, Z_2 for $X^\perp \subset T_p M$. Another, apparently more principled, way of computing this average is

$$\frac{1}{2\pi} \int_0^{2\pi} K(X, (\cos \theta)Z_1 + (\sin \theta)Z_2) d\theta.$$

However, by multilinearity of the curvature tensor R and the Riemannian metric g , the above integral is equal to

$$\begin{aligned}
& \frac{1}{2\pi} \int_0^{2\pi} g(R(X, (\cos \theta)Z_1 + (\sin \theta)Z_2)((\cos \theta)Z_1 + (\sin \theta)Z_2), X) d\theta \\
&= \frac{1}{2\pi} \int_0^{2\pi} [\cos^2 \theta g(R(X, Z_1)Z_1, X) + \sin \theta \cos \theta g(R(X, Z_1)Z_2, X) \\
&\quad + \sin \theta \cos \theta g(R(X, Z_2)Z_1, X) + \sin^2 \theta g(R(X, Z_2)Z_2, X)] d\theta \\
&= \frac{1}{2\pi} \left[K(X, Z_1) \int_0^2 \pi \cos^2 \theta d\theta + (g(R(X, Z_1)Z_2, X) + g(R(X, Z_2)Z_1, X)) \int_0^2 \pi \sin \theta \cos \theta d\theta \right. \\
&\quad \left. + K(X, Z_2) \int_0^{2\pi} \sin^2 \theta d\theta \right] \\
&= \frac{1}{2\pi} \int_0^{2\pi} K(X, (\cos \theta)Z_1 + (\sin \theta)Z_2) d\theta,
\end{aligned}$$

so we get the same thing either way we compute.

Exercise 4.9.2. Generalize [Example 4.9.1](#) as follows. Let M be a Riemannian n -manifold, $p \in M$, and Z_1, \dots, Z_{n-1}, X an orthonormal basis for $T_p M$. Again, we want to take the average of sectional curvatures over all 2-planes containing X , and there are two ways to do this. Show that both computations yield the same answer.

In other words, it's enough just to average over any orthonormal basis and this will give the same answer as averaging over every 2-plane containing X . In fact, for reasons that will become apparent when we discuss this a little more abstractly, most people don't actually take averages, they just sum over orthonormal bases:

Definition 4.9.3. Let M be an n -dimensional Riemannian manifold, $p \in M$ and $X \in T_p M$ a unit vector. Complete X to an orthonormal basis Z_1, \dots, Z_{n-1}, X for $T_p M$. The *Ricci curvature* of M at p in the direction of X is

$$\text{Ric}_p(X) = \sum_{i=1}^{n-1} K(X, Z_i).$$

By [Exercise 4.9.2](#), $\frac{1}{n-1} \text{Ric}_p(X)$ is the average (in *either* sense) of the sectional curvatures of all 2-planes containing X . Of course, there's no reason to stop averaging there. What if we take the average of Ricci curvatures over all directions? Again, we're going to sum rather than averaging, but this gives scalar curvature:

Definition 4.9.4. Let M be a Riemannian manifold, $p \in M$, and Z_1, \dots, Z_n an orthonormal basis for $T_p M$. The *scalar curvature* of M at p is

$$\text{Scal}(p) := \sum_{i=1}^n \text{Ric}_p(Z_i) = \sum_{i,j} K(Z_i, Z_j).$$

Exercise 4.9.5. Prove that $\frac{1}{n} \text{Scal}(p)$ is the average over all unit vectors X in $T_p M$ of the Ricci curvatures $\text{Ric}_p(X)$.

Exercise 4.9.6. Prove that $\frac{1}{n(n-1)}$ is the average over all 2-dimensional subspaces $\sigma \subset T_p M$ of the sectional curvatures $K(\sigma)$. The key issue here is to figure out what it means to average over 2-dimensional subspaces.

Example 4.9.7. If M is n -dimensional and has constant sectional curvature K , then it also has constant Ricci curvature $\text{Ric}_p(X) = (n-1)K$ and constant scalar curvature $\text{Scal}(p) = n(n-1)K$. For example, S^3 has Ricci curvature equal to 2 and scalar curvature equal to 6 everywhere.

Example 4.9.8. Recall from Example 4.8.9 our orthonormal basis $\{H_1, H_2, \frac{1}{\sqrt{2}}E, \frac{1}{\sqrt{2}}F, \frac{1}{\sqrt{2}}X, \frac{1}{\sqrt{2}}Y\}$ for the left-invariant vector fields on $\text{SO}(4)$. If U is any of these basis vectors, then we can compute $\text{Ric}_h(U)$ (at any $h \in \text{SO}(4)$) by summing the appropriate row of Table 4.2. But all row sums are equal to 1, so we see that, despite not having constant sectional curvature, $\text{SO}(4)$ has constant Ricci curvature equal to 1 everywhere—and hence, since $\text{SO}(4)$ is 6-dimensional, constant scalar curvature equal to 6 everywhere.

Now we make things a bit more abstract. By definition, if $X \in T_p M$ is a unit vector, then we can complete to an orthonormal basis $Z_1, \dots, Z_{n-1}, Z_n = X$ in any way we like and we have

$$\text{Ric}_p(X) = \sum_{i=1}^{n-1} K(X, Z_i).$$

Writing $K(X, Z_i) = g(R(X, Z_i)Z_i, X)$ (since $\|X \wedge Z_i\|^2 = 1$ for all i), we can also write this as

$$\text{Ric}_p(X) = \sum_{i=1}^{n-1} g(R(X, Z_i)Z_i, X).$$

In fact, since skew-symmetry of R implies that $R(X, Z_n) = R(X, X) = 0$, we can sum from 1 to n instead:

$$\text{Ric}_p(X) = \sum_{i=1}^n g(R(X, Z_i)Z_i, X) = \sum_{i=1}^n g(R(Z_i, X)X, Z_i). \quad (4.33)$$

In other words, $\text{Ric}_p(X)$ is the trace of the linear map $T_p M \rightarrow T_p M$ given by

$$U \mapsto R(U, X)X.$$

In particular, since trace is the sum of the diagonal entries in *any* matrix representation of a linear transformation, this shows that the formula (4.33) for $\text{Ric}_p(X)$ is valid for *any* orthonormal basis Z_1, \dots, Z_n for $T_p M$: Z_n is not required to equal X .

Now, if we forget about the requirement that X be a unit vector, the expression

$$\sum_{i=1}^n g(R(Z_i, X)X, Z_i)$$

is a quadratic form, which we can polarize to get a symmetric bilinear form

$$\text{Ric}_p(X, Y) := \sum_{i=1}^n g(R(Z_i, X)Y, Z_i),$$

which we've written here in terms of an orthonormal basis. Again, this is a trace so we can write more abstractly.

Definition 4.9.9. Let M be a Riemannian manifold. The *Ricci tensor* is a symmetric $(0, 2)$ -tensor field on M which is defined at each $p \in M$ as the map $\text{Ric}_p : T_p M \times T_p M \rightarrow \mathbb{R}$ given by

$$\text{Ric}_p(X, Y) := \text{tr}(U \mapsto R(U, X)Y).$$

Let's recall how to compute the trace of a linear transformation of $T_p M$. Say X_1, \dots, X_n is a basis for $T_p M$, and as usual write $g_{ij} := g_p(X_i, X_j)$. If $A : T_p M \rightarrow T_p M$ is linear, with matrix $[a_j^i]_{i,j}$ with respect to our chosen basis, then that means

$$A(X_j) = \sum_i a_j^i X_i,$$

so we have

$$g_p(A(X_j), X_k) = g_p\left(\sum_i a_j^i X_i, X_k\right) = \sum_i a_j^i g_p(X_i, X_k) = \sum_i a_j^i g_{ik}.$$

Therefore, if the inverse matrix of $[g_{ij}]_{i,j}$ is $[g^{k\ell}]_{k,\ell}$, then we have

$$\text{tr}(A) = \sum_i a_i^i = \sum_{i,k,\ell} a_k^i g_{i\ell} g^{\ell k} = \sum_{k,\ell} g_p(A(X_k), X_\ell) g^{\ell k}.$$

In particular, then, if $X_i = \frac{\partial}{\partial x_i}$ in some local coordinate chart around p , then this implies that

$$\text{Ric}_{ij}(p) = \text{Ric}_p(X_i, X_j) = \text{tr}(U \mapsto R(U, X_i)X_j) = \sum_{k,\ell} g_p(R(X_k, X_i)X_j, X_\ell) g^{\ell k} = \sum_{k,\ell} R_{j\ell ki} g^{\ell k} = \sum_k R_{jki}^k.$$

This is symmetric, so we can also write $\text{Ric}_{ij}(p) = \sum_k R_{ikj}^k$ or, in abstract index notation (with the Einstein summation convention),

$$\text{Ric}_{ij} = R_{ikj}^k = g^{k\ell} R_{k\ell ij};$$

that is, the Ricci tensor is the contraction of the Riemann curvature tensor in the first and third indices.

Exercise 4.9.10. Show that the scalar curvature is the trace (or contraction) of the Ricci tensor.

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