How Powerful is Adiabatic Quantum Computation?

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Abstract:
We analyze the computational power and limitations of the recently proposed 'quantum adiabatic evolution algorithm'.

1   Introduction

Quantum computation is a revolutionary idea that has fundamentally transformed our notion of feasible computation. The most dramatic example of the power of quantum algorithms was exhibited in Shor's celebrated quantum algorithms for factoring and discrete log [13]. Grover's quantum search algorithm [10] gives a quadratic speedup for a much wider class of computational problems. Despite numerous attempts in the last few years, it has proved to be a difficult challenge to design new quantum algorithms. Recently, Farhi et al. [6, 7] proposed a novel paradigm for the design of quantum algorithms — via quantum adiabatic evolution. This paradigm bears some resemblance to simulated annealing, in the sense that the algorithm starts from an initial disordered state, and homes in on a solution (by what could be described as quantum local search) as a parameter 's' is smoothly varied from 0 to 1. The challenge lies in showing that the process still converges to the desired solution with non-negligible probability if this transition is made in polynomial time. In [7, 8], this paradigm was applied to the Exact Cover problem (which has a close connection to the 3SAT problem), and using computer simulations it was shown that the algorithm works efficiently on small randomly chosen instances of this problem.

In the first part of the article, we discuss the quantum adiabatic theorem and explain the quantum adiabatic approach to computation. Next, we clarify the connection between the continuous time evolution of adiabatic computing and the quantum circuit model with its discretized time. We do this by describing a way of efficiently simulating quantum adiabatic algorithms with a network of standard quantum gates. After this exposition, we explore three questions about quantum adiabatic evolution algorithms.

Can we apply the exponential lower bounds for quantum search [2] to conclude that the adiabatic quantum algorithm for 3SAT must take exponential time? More concretely, at a high level of abstraction, the adiabatic quantum algorithm for 3SAT may be viewed as some quantum process that gets information about the 3SAT instance only by (quantum) queries of the following type: given a truth assignment, how many clauses of the formula Φ are not satisfied? We prove that there is a (classical) polynomial time algorithm that can reconstruct the 3CNF formula Φ by making polynomially many queries of this type. It is somewhat surprising that this question does not appear to have been studied in the context of relativization results for NP. In our context, it rules out any query complexity based (quantum) lower bound for the adiabatic quantum solution of 3SAT.

Is adiabatic quantum computing really quantum? We give an example of an adiabatic quantum algorithm for searching that matches the optimal quadratic speedup obtained by Grover's search algorithm. This example demonstrates that the 'quantum local search', which is implicit in the adiabatic evolution, is truly non-classical in nature from a computational viewpoint.

Finally, we give a simple example of a computational problem on which the adiabatic quantum algorithm provably takes exponential time. Although the problem is easy to solve classically, it is designed to be difficult for algorithms

based on local search: its global optimum lies in a narrow basin, while there is a local optimum with a much larger basin. Let f be a function on the n-bit strings, where $f(x)$ depends only on $w(x)$, the Hamming weight of x. The problem is to find an x that minimizes $f(x)$. (Obviously, it is straightforward to solve this class of problems in $n + 1$ steps.) Consider functions f such that for $w(x) \leq (\frac{1}{2} + \varepsilon)n$, $f(x) = w(x)$, and which decreases for $w(x) > (\frac{1}{2} + \varepsilon)n$ to the global minimum $f(1^n) = -1$. We prove that for such instances, the adiabatic quantum algorithm requires an exponential slowdown in n. We do this by showing that the gap between the minimum and second eigenvalue of the Hamiltonian of the system is exponentially small. In an upcoming paper [5], we generalize these techniques to show a similar exponential slowdown for 3SAT.

2   The Quantum Adiabatic Theorem

The Hamiltonian of a physical system gives a complete specification of the time evolution of this system. At a given time t, let $\psi(t)$ denote the state of the system under the influence of the Hamiltonian $H(t)$. The differential equation that describes the time evolution is the well-known Schrödinger equation:

$$i\hbar \tfrac{d}{dt}|\psi(t)\rangle \;\; = \;\; H(t)|\psi(t)\rangle,$$

where $\hbar$ is Planck's constant $h \approx 6.63 \times 10^{-34}$ Joule-second, divided by 2π. A Hamiltonian is described by a Hermitian matrix, whose eigenvectors represent the eigenstates of the system. The corresponding eigenvalues refer to the different energies of the eigenstates. The state (eigenvector) with the lowest energy (eigenvalue) is called the 'ground state' of the system. The Schrödinger equation can also be described with reference to the unitary transformation U that is defined by the Hamiltonian $H(t)$ (from now on we work with $\hbar = 1$):

$$\tfrac{d}{dt}U(t) \;\; = \;\; -iH(t)U(t),$$

with the initial condition $U(0) = I$. We say that the Hamiltonian evolution from $H(0)$ to $H(T)$ *induces* the unitary transformation $U(T)$. The evolution of a system with a time-independent Hamiltonian H is easily expressed by the exponential $U(T) = e^{-iTH}$. Finding the (approximate) solutions for Hamiltonians that vary in time is one of the core tasks in quantum physics. One of the most important cases of such a time-dependent case is described by the adiabatic evolution of an isolated quantum mechanical system.

The quantum adiabatic theorem states that a physical system that is initially in its ground state, tends to stay in this lowest energy state, provided that the Hamiltonian of the system is changed 'slowly enough'.[4]

The quantitative version of the adiabatic theorem gives the following specific upper bound on the slowdown that is required for the adiabatic evolution of the ground state. (See for example [12] for more details on this.) Parameterize the time-dependent Hamiltonian by $H(s)$ for $0 \leq s \leq 1$ and its ground state by $\phi(s)$. Our goal is thus to gradually transform the applied Hamiltonian from $H(0)$ to $H(1)$ such that the initial state $\psi(0) = \phi(0)$ evolves to a close approximation $\psi(1) \approx \phi(1)$ of the ground state of $H(1)$. We introduce a delay factor $\tau(s)$, which determines the rate at which the Hamiltonian is modified as a function of s. Now the Schrödinger equation in s equals

$$\tfrac{d}{ds}|\psi(s)\rangle \;\; = \;\; -i\tau(s)H(s)|\psi(s)\rangle.$$

The crucial quantity for this transformation is the gap between the two smallest eigenvalues of $H(s)$, which we denote by $g(s)$. It can be shown that a delay schedule τ with

$$\tau(s) \;\; \gg \;\; \frac{\|\frac{d}{ds}H(s)\|_2}{g(s)^2}$$

is 'sufficiently slow' for the adiabatic evolution from $\phi(0)$ to $\phi(1)$. As a result, the total delay of this process will be of the order $\int_{s=0}^1 \tau(s)ds$. For most Hamiltonians it is too difficult to determine the gap $g(s)$ for every s. If this is the case, we can also look at the *minimum gap* $g_{\min} := \min_s g(s)$ and the maximum $\Delta_{\max} := \max_s \|\frac{d}{ds}H(s)\|_2$, and obtain the adiabatic evolution with the constant delay factor $\tau(s) = \tau_c \in O(\frac{\Delta_{\max}}{g_{\min}^2})$.

3 Adiabatic Quantum Computation

Adiabatic quantum computation, as proposed by Farhi et al.[6], works as follows. At time $t = 0$, the quantum mechanical system is described by a Hamiltonian H_0, whose eigenstates are easy to compute. Next, this system is slowly transformed to its final Hamiltonian H_f, for which the ground state is the solution to a specific minimization problem f. We do this is by letting the energies λ_z of the eigenstates z of H_f correspond with the function that we try to minimize. Hence, if this function f has domain $\{0,1\}^n$, then the final Hamiltonian is defined by

$$H_f \quad := \quad \sum_{z \in \{0,1\}^n} f(z) \cdot |z\rangle\langle z|.$$

We will assume throughout this paper that $f : \{0,1\}^* \to \mathbb{R}$ is computable in polynomial time, and that $f(x)$ is bounded by a polynomial in $|x|$.

The choice of the initial Hamiltonian H_0 is independent of the solution of the problem, and will be such that H_0 is not diagonal in the computational z-basis. Specifically, we consider the 'Hadamard basis' with the bit values

$$|\hat{0}\rangle := \tfrac{1}{\sqrt{2}}(|0\rangle + |1\rangle) \quad \text{and} \quad |\hat{1}\rangle := \tfrac{1}{\sqrt{2}}(|0\rangle - |1\rangle).$$

For a binary string $z \in \{0,1\}^n$, let $|\hat{z}\rangle$ denote the state which would be written as $|z\rangle$ in this basis. (The unitary mapping between these two representations is provided by the n-fold Hadamard matrix: $W^{\otimes n}|z\rangle = |\hat{z}\rangle$ and $W^{\otimes n}|\hat{z}\rangle = |z\rangle$.)

A simple starting Hamiltonian that fulfills the above requirements is

$$H_0 \quad := \quad \sum_{z \in \{0,1\}^n} h(z) \cdot |\hat{z}\rangle\langle\hat{z}|,$$

with $h(0^n) = 0$ and $h(z) \geq 1$ for all other $z \neq 0^n$, such that the ground state with zero energy of H_0 is the uniform superposition $|\hat{0}\cdots\hat{0}\rangle = \frac{1}{\sqrt{2^n}} \sum_z |z\rangle$. Having defined the initial and final conditions of our system, we will now describe the time-evolution.

Following the proposal by Farhi et al. in [6, 7], we can define the time dependent Hamiltonian $H(t)$ as the linear combination of the starting and the final Hamiltonian:

$$H(t) \quad := \quad \left(1 - \tfrac{t}{T}\right) H_0 + \tfrac{t}{T} H_f,$$

with $0 \leq t \leq T$, and T the crucial *delay factor* of the $H_0 \to H_f$ transition.

By the adiabatic theorem we know that this system will map the initial ground state $|\psi(0)\rangle = |\hat{0}^n\rangle$ to the global minimum of the function f, provided that we pick T large enough. In the previous section we mentioned that $T \in O(\Delta_{\max} g_{\min}^{-2})$ is a sufficient upper bound on this delay. Without any further knowledge about the specific Hamiltonian $H(t)$ — which involves detailed knowledge about the function f, this is also a lower bound for a reliable adiabatic evolution from H_0 to H_f. Because $\|\frac{d}{ds}H(s)\|_2$ is polynomial in n (as long as $f \in \text{poly}(n)$), we will ignore this factor and focus mostly on the $T \gg g_{\min}^{-2}$ requirement for the delay of the adiabatic quantum computation.

4 Approximating the Adiabatic Evolution

In this section we explain how the continuous time evolution from H_0 to H_f can be approximated by a quantum circuit of size $\text{poly}(nT)$. Our goal is to demonstrate the ingredients of the polynomial upper bound, and we do not try to optimize to get the most efficient simulation.

The approximation is established in two steps. First, we discretize the evolution from H_0 to H_f by a finite sequence of Hamiltonians $H_1', H_2', \ldots$ that gives rise to the same overall behavior. Second, we show how at any moment the combined Hamiltonian $H_j' = (1-s)H_0 + sH_f$ can be approximated by interleaving two simple unitary transformations.

To express the error of our approximation, we use the ℓ_2 induced operator norm "$\| \circ \|_2$":

$$\|M\|_2 \quad := \quad \max_{\|x\|_2=1} \|Mx\|_2.$$

The next lemma compares two Hamiltonians $H(t)$ and $H'(t)$ and their respective unitary transformations $U(T)$ and $U'(T)$.

Lemma 1 *Let $H(t)$ and $H'(t)$ be two time-dependent Hamiltonians for $0 \leq t \leq T$, and let $U(T)$ and $U'(T)$ be the respective unitary evolutions that they induce. If the difference between the Hamiltonians is limited by $\|H(t) - H'(t)\|_2 \leq \delta$ for every t, then the distance between the induced transformations is bounded by $\|U(T) - U'(T)\|_2 \leq \sqrt{2T\delta}$.*

Proof: Let $\psi(t)$ and $\psi'(t)$ be the two state trajectories of the two Hamiltonians H and H' with initially $\psi(0) = \psi'(0)$. Then, for the inner product between the two states (with initially $\langle \psi'(0)|\psi(0)\rangle = 1$), we have

$$\tfrac{d}{dt}\langle \psi'(t)|\psi(t)\rangle = -\mathrm{i}\langle \psi'(t)|(H(t) - H'(t))|\psi(t)\rangle.$$

Because at any moment t we have $\||\psi(t)\rangle\|_2 = \||\psi'(t)\rangle\|_2 = 1$ and $\|H(t) - H'(t)\|_2 \leq \delta$, we see that at $t = T$ the lower bound $|\langle \psi'(T)|\psi(T)\rangle| \geq 1 - T\delta$ holds. This confirms that for every vector ψ we have $\|U(T)|\psi\rangle - U'(T)|\psi\rangle\|_2 \leq \sqrt{2T\delta}$. ∎

This lemma tells us how we can deviate from the ideal Hamiltonian $H(t) := (1 - \frac{t}{T})H_0 + \frac{t}{T}H_f$, without introducing too big of an error to the induced evolution. As mentioned above, we will approximate the continuous $H(0) \to H(T)$ trajectory by a sequence of r Hamiltonians $H'_1, \ldots, H'_r$, each of which applied for a duration of $\frac{T}{r}$. This yields the unitary evolution $U'(T)$, defined by

$$U'(T) := \mathrm{e}^{-\mathrm{i}(\frac{T}{r})H'_r} \cdots \mathrm{e}^{-\mathrm{i}(\frac{T}{r})H'_1},$$

with for any $1 \leq j \leq r$ the Hamiltonian $H'_j := H(\frac{jT}{r}) = (1 - \frac{j}{r})H_0 + (\frac{j}{r})H_f$. If we view H' as a time-dependent Hamiltonian $H'(t) := H_{j(t)}$ with $j(t) := \lceil \frac{rt}{T} \rceil$, then we have the bound $\|H(t) - H'(t)\|_2 \leq \frac{1}{r}\|H_f - H_0\|_2 \in O(n^d/r)$ for all t. By the previous lemma we thus have the bound $\|U(T) - U'(T)\|_2 \in O(\sqrt{Tn^d/r})$.

The second part of our approximation deals with the problem of implementing the unitary transformations U'_j defined by

$$U'_j := \mathrm{e}^{-\mathrm{i}\frac{T}{r}(1 - \frac{j}{r})H_0 - \mathrm{i}\frac{T}{r}(\frac{j}{r})H_f}.$$

with elementary operations.

The Campbell-Baker-Hausdorff theorem[3] tells us how well we can approximate 'parallel Hamiltonians' by consecutive ones: $\|\mathrm{e}^{A+B} - \mathrm{e}^A\mathrm{e}^B\|_2 \in O(\|AB\|_2)$. Hence in our case, by defining

$$U''_j := \mathrm{e}^{-\mathrm{i}\frac{T}{r}(1 - \frac{j}{r})H_0} \cdot \mathrm{e}^{-\mathrm{i}\frac{T}{r}(\frac{j}{r})H_f},$$

we get the approximation $\|U'_j - U''_j\|_2 \in O(\frac{T^2}{r^2}\|H_0 H_f\|_2)$. This leads to $\|U'(T) - U''(T)\|_2 \in O(n^{d+1}T^2/r)$, and hence also for the original transformation: $\|U(T) - U''(T)\|_2 \in O(n^{d+1}T^2/r)$.

Because $H_0 = \sum_z h(z)|\hat{z}\rangle\langle\hat{z}|$ is diagonal in the Hadamard basis $\{\hat{0}, \hat{1}\}^n$, and $H_f = \sum_z f(z)|z\rangle\langle z|$ is diagonal in the computational bases, we can implement the above U''_j as

$$U''_j = W^{\otimes n} \cdot F_{0,j} \cdot W^{\otimes n} \cdot F_{f,j},$$

with $W^{\otimes n}$ the n-fold Hadamard transform, and F_0 and F_f the appropriate phase changing operations:

$$F_{0,j}|z\rangle := \mathrm{e}^{-\mathrm{i}\frac{T}{r}(1 - \frac{j}{r})h(z)}|z\rangle,$$
$$F_{f,j}|z\rangle := \mathrm{e}^{-\mathrm{i}\frac{T}{r}(\frac{j}{r})f(z)}|z\rangle.$$

Because $h(z)$ and $f(z)$ are easy to compute, so are F_0 and F_f. We have thus obtained the following theorem.

Theorem 1 *Let H_0 and H_f be the initial and final Hamiltonians used in an adiabatic computation, with the function $f \in O(n^d)$. Then, the unitary transformation $U(T)$ induced by the time-dependent Hamiltonian $H(t) := (1 - \frac{t}{T})H_0 + \frac{t}{T}H_f$ can be approximated by r consecutive unitary transformations $U''_1, \ldots, U''_r$ with $r \in O(T^2 n^{d+1})$. Furthermore, each U''_j has the form $W^{\otimes n}F_0 W^{\otimes n}F_f$ and can thus be efficiently implemented in $\mathrm{poly}(nT)$ time.*

It is interesting to note that the $W^{\otimes n}F_0 W^{\otimes n}F_f$ transformation has the same form as the 'Grover iteration' of the standard quantum search algorithm[10]. More recently, we also learned that the work of Hogg on quantum search heuristics[11] describes essentially the same algorithm as the adiabatic approach to minimization.

5 Quantum Adiabatic Searching

One question that should be asked first is if adiabatic quantum computing is truly quantum computing. In this section we answer this question affirmatively by reproducing the quadratic speed-up of Lov Grover's search algorithm.

For the search problem, the function $f : \{0,1\}^n \to \mathbb{R}$ takes on value 1 on all strings except the solution $u \in \{0,1\}^n$ for which $f(u) = 0$. Thus the final Hamiltonian for the adiabatic algorithm, H_u, will have eigenstates $|z\rangle$ with eigenvalue 1, with the exception of the unknown solution $u \in \{0,1\}^n$, which has eigenvalue 0:

$$H_u \quad := \quad \sum_{z \in \{0,1\}^n \setminus \{u\}} |z\rangle\langle z|.$$

The initial Hamiltonian is defined similarly, except that it is diagonal in the Hadamard basis, and has ground state $|\hat{0}^n\rangle$:

$$H_0 \quad := \quad \sum_{z \in \{0,1\}^n \setminus \{0^n\}} |\hat{z}\rangle\langle \hat{z}|.$$

With these initial and final conditions one can easily show that for the resulting time-dependent Hamiltonian

$$H(t) \quad := \quad (1 - \tfrac{t}{T})H_0 + \tfrac{t}{T}H_u,$$

the gap between the two smallest eigenvalues as a function of $s := \frac{t}{T}$ is expressed by

$$g(s) \quad = \quad \sqrt{\frac{2^n + 4(2^n - 1)(s^2 - s)}{2^n}}. \tag{1}$$

This gap reaches its minimum at $t = \frac{T}{2}$ when it equals $\frac{1}{\sqrt{2^n}}$. At first sight, this would lead to the conclusion that the necessary delay factor $T = \Omega(g_{\min}^{-2})$ is linear in $N = 2^n$. However, by using our knowledge of the gap function $g(\frac{t}{T})$ we can significantly reduce the running time to $O(\sqrt{N})$.

For example, regardless of the solution u, we know that the transition from $H(0)$ to $H(\frac{T}{3})$ will have a minimal gap that is significantly bigger than $\frac{1}{\sqrt{N}}$. The necessary delay factor that we use for this first part of our transformation $H_0 \to H_u$, can therefore be much smaller than N. In general at any moment $s = \frac{t}{T}$, Equation 1 tells us the size of the gap $g(s)$, and hence the delay factor that suffices at that moment. This means that we can employ a varying delay factor $g(s)^{-2}$, without destroying the desired adiabatic properties of the evolution $H_0 \to H_u$. In sum, this approach leads to a total delay factor of

$$\int_{s=0}^{1} \frac{ds}{g(s)^2} \quad = \quad \int_{s=0}^{1} \frac{2^n}{2^n + 4(2^n - 1)(s^2 - s)} ds$$

$$= \quad \frac{2^n \cdot \arctan(\sqrt{2^n - 1})}{\sqrt{2^n - 1}}.$$

As a function of $N = 2^n$, this gives a time complexity $O(\sqrt{2^n}) = O(\sqrt{N})$, which coincides with the well-known square root speed-up of quantum searching. (See the article by Farhi and Gutmann[9] for another example of a 'continuous time algorithm' for quantum searching.)

6 Query Bounds for the 3SAT Problem

The adiabatic quantum algorithms of [7, 8] work on 3SAT as follows: on input a formula $\Phi = C_1 \wedge \cdots \wedge C_M$ (where the C_i are clauses in variables $x_1, \ldots, x_n$), the only way the quantum algorithm gathers information about Φ is by queries which ask, for a given truth assignment b (in general a superposition of assignments), how many of the M clauses b does not satisfy. A natural approach to establishing a lower bound on the running time of the adiabatic quantum algorithm is to show that any quantum algorithm must make a large number of such queries to solve the problem. This is the approach that leads to the exponential lower bound for unstructured search [2] (there the query asked, for a given assignment b, whether or not it is a satisfying assignment), thus showing that relative to a random

oracle NP is not a subset of subexponential quantum time. In this section, we show that the seemingly small difference between the specifications of these two types of queries results in a dramatic change in the query complexity — $O(n^3)$ queries suffice to obtain enough information to characterize Φ. Thus black box or oracle techniques do not rule out a polynomial time solution to 3SAT by adiabatic quantum search. To reconcile this with the oracle results from [2], it is useful to recall that the Cook-Levin theorem, suitably formulated as saying that NP has a 'local-checkability' property, does not relativize [1] (see [14] for a brief discussion of this issue). In this sense, the results in this section indicate that even keeping track about the number of unsatisfied clauses constitutes sufficient structural information about the problem to bypass the oracle results.

More formally, let

$$F_\Phi(b) \quad := \quad \text{``\# unsatisfied clauses in assignment } \Phi(b)\text{''},$$

with $b \in \{0,1\}^n$. In our black box model, the quantum algorithm is only allowed to access Φ via a quantum black-box B_Φ that reversibly maps $|b\rangle|0\rangle \mapsto |b\rangle|F_\Phi(b)\rangle$. In this section, we prove that the query complexity for 3SAT is $O(n^3)$, by showing that F_Φ is completely determined by its values on the $O(n^3)$ input strings of Hamming weight ≤ 3. Our techniques also apply to the Exact Cover problem discussed in [7].

For convenience, and without loss of generality, we will not allow repeated variables in the same clause, but instead will allow clauses of size less than 3. For example, we can replace the clause $(x_1 \vee x_1 \vee x_2)$ with $(x_1 \vee x_2)$, and $(x_1 \vee \neg x_1 \vee x_2)$ with a constant clause (1) that is always satisfied. Without loss of generality, we can assume that the number of such (1) clauses is 0.

Let us introduce some notation. Let $|XXX|$ denote the number of clauses in Φ that have all three variables without negation (e.g. $(x_1 \vee x_2 \vee x_3)$). We will say that these clauses are "of the form" XXX. Let $|\overline{X}XX|$ denote the number of clauses that have exactly one variable negated (e.g. $(x_1 \vee \neg x_2 \vee x_3)$). Further, we let $|\overline{X}\overline{X}X|$ denote the number of clauses that have exactly two variable negated, and $|\overline{X}\overline{X}\overline{X}|$ denote the number of clauses that have all three variables negated. We also define the analogous 1 and 2 variable versions of these expressions.

Furthermore, if we subscript any of the X with an index, say i, then we only count clauses that have x_i as one of the non-negated (or *positive*) variables. Similarly, if we subscript any of the $\overline{X}$ with an index, say i, then we only count clauses that have $\overline{x_i}$ as one of the negated variables. For example, $|X_i XX|$ denotes the number of clauses in Φ that contain the variable x_i and two other positive variables, $|\overline{X}_i X_j X|$ denotes the number of clauses with $\overline{x_i}$ and x_j and another positive variable, and $|X_i X_j X|$ denotes the number of clauses that have one of the positive variables equal to x_i, another equal to x_j, and another positive variable. The expression $|\overline{X}_i \overline{X}_j X_k|$ equals the number of times the clause $(\neg x_i \vee \neg x_j \vee x_k)$ (or equivalent permuted clauses like $(x_k \vee \neg x_j \vee \neg x_i)$) occurs.

These expressions are symmetric under permutation of the symbols, so for example, $|X_i X_j X| = |X_j X_i X|$ and $|\overline{X}_i X_j X_k| = |\overline{X}_i X_k X_j|$.

For example, we have that

$$F_\Phi(0^n) \quad = \quad |XXX| + |XX| + |X|$$

since any clause with a negated variable will be satisfied, and the rest will not be satisfied.

The following definitions will be helpful. For each $i \in \{1, \ldots, n\}$ let

$$Y_i \quad := \quad |\overline{X}_i XX| - |X_i XX| + |\overline{X}_i X| - |X_i X| + |\overline{X}_i| - |X_i|.$$

For each pair $i, j \in \{1, \ldots, n\}$, $i \neq j$, let

$$Y_{ij} \quad := \quad |\overline{X}_i X_j X| + |X_i X_j X| - |\overline{X}_j X_i X| - |\overline{X}_i X_j X| + |\overline{X}_i X_j| + |X_i X_j| - |\overline{X}_i X_j| - |\overline{X}_j X_i|.$$

For each triple i, j, k of pairwise distinct integers from $\{1, \ldots, n\}$, let

$$Y_{ijk} \quad := \quad |\overline{X}_k X_i X_j| + |\overline{X}_i X_j X_k| + |\overline{X}_j X_i X_k| + |\overline{X}_i X_j X_k| - |X_i X_j X_k| - |\overline{X}_j X_i X_k| - |\overline{X}_i X_k X_j| - |\overline{X}_j X_k X_i|.$$

For each $i \in \{1, \ldots, n\}$ let e^i denote the string with a 1 in the ith position and 0s elsewhere. For each $i, j \in \{1, \ldots, n\}$, $i \neq j$, let e^{ij} denote the string with a 1 in positions i and j and 0s elsewhere. For each $i, j, k \in \{1, \ldots, n\}$, pairwise distinct, let e^{ijk} denote the string with a 1 in positions i, j and k and 0s elsewhere.

We now have the next theorem.

Theorem 2 *Let $b \in \{0,1\}^n$ and let I be the subset of $\{1,\ldots,n\}$ such that $b_i = 1 \iff i \in I$. Then*

$$F_\Phi(b) \quad = \quad F_\Phi(0^n) + \sum_{i \in I} Y_i + \sum_{i < j \in I} Y_{ij} + \sum_{i < j < k \in I} Y_{ijk}.$$

Furthermore,

$$
\begin{aligned}
Y_i &= F_\Phi(e^i) - F_\Phi(0^n) \\
Y_{ij} &= F_\Phi(e^{ij}) - F_\Phi(e^i) - F_\Phi(e^j) + F_\Phi(0^n) \\
Y_{ijk} &= F_\Phi(e^{ijk}) - F_\Phi(e^{ij}) - F_\Phi(e^{ik}) - F_\Phi(e^{jk}) + F_\Phi(e^i) + F_\Phi(e^j) + F_\Phi(e^k) - F_\Phi(0^n).
\end{aligned}
$$

In other words, in order to be able to evaluate F_Φ for every input string $\{0,1\}^n$, we only need to query the black-box B_Φ on the $O(n^3)$ inputs with Hamming weight at most 3 (the cases $b \in \{0^n, e^i, e^{ij}, e^{ijk}\}$). Specifically, we can decide whether Φ is satisfiable or not by querying the black-box B_Φ a total of $O(n^3)$ times, after which we use the query results to evaluate F_Φ for all other possible inputs $b \in \{0,1\}^n$. If any of the strings give $F_\Phi(b) = 0$, then Φ is satisfiable, otherwise it is not satisfiable. (Clearly, with this information we can also answer other decision problems like "$\Phi \in \text{PP}$?") The full proof of this theorem is described in the appendix of this article.

7 Lower Bounds for Adiabatic Algorithms

In this section we present an easy n-bit problem, for which the adiabatic approach only succeeds if it is allowed an exponential delay. We do this by changing an easy problem (the Minimum Hamming Weight Problem) into a perturbed version for which the proper solution is as far as possible from its local minimum. It will be shown that for this perturbed version, the quantum adiabatic algorithm does indeed require exponential time.

7.1 The Minimum Hamming Weight Problem

Consider the adiabatic quantum algorithm that tries to minimize the Hamming weight $w(z)$ of an n bit string $z \in \{0,1\}^n$. We define the initial Hamiltonian by $H_0 := \sum_z w(z)|\hat{z}\rangle\langle\hat{z}|$, such that the time-dependent Hamiltonian is

$$H_w(t) \quad := \quad \left(1 - \tfrac{t}{T}\right) \sum_{z \in \{0,1\}^n} w(z)|\hat{z}\rangle\langle\hat{z}| + \tfrac{t}{T} \sum_{z \in \{0,1\}^n} w(z)|z\rangle\langle z|.$$

As intended, the ground state of the final Hamiltonian is simply $|0\cdots0\rangle$ with zero energy.

Since $w(z) = z_1 + \cdots + z_n$, it is easy to see that $H_w(t)$ is a sum of n Hamiltonians, each acting on a single qubit. Thus even though $H_w(t)$ is a $2^n \times 2^n$ dimensional matrix, which thus has 2^n eigenstates, these eigenstates and their corresponding eigenvalues may be computed by solving the 2 dimensional problem. For the analysis of the minimal gap between the two smallest eigenvalues it is again convenient to introduce a relative time-parameter $s := \tfrac{t}{T}$, which ranges from 0 to 1. The eigen-decomposition for the 2 dimensional problem yields:

$$\frac{1}{2}\begin{pmatrix} 1-s & s-1 \\ s-1 & 1+s \end{pmatrix} \quad = \quad \lambda_0(s)|v_0(s)\rangle\langle v_0(s)| + \lambda_1(s)|v_1(s)\rangle\langle v_1(s)|. \tag{2}$$

with

$$\lambda_0(s) = \tfrac{1}{2} - \tfrac{1}{2}\sqrt{2s^2 - 2s + 1} \quad \text{and} \quad \lambda_1(s) = \tfrac{1}{2} + \tfrac{1}{2}\sqrt{2s^2 - 2s + 1}.$$

Specifically, at $s = 0$ we have $|v_0(0)\rangle = |\hat{0}\rangle = \tfrac{1}{\sqrt{2}}(|0\rangle + |1\rangle)$ and $|v_1(0)\rangle = |\hat{1}\rangle = \tfrac{1}{\sqrt{2}}(|0\rangle - |1\rangle)$, while at $s = 1$ we have $|v_0(1)\rangle = |0\rangle$ and $|v_1(1)\rangle = |1\rangle$.

For the n qubit case, it is easily shown that for every $y \in \{0,1\}^n$ there is an eigenvalue

$$\lambda_y(s) \quad = \quad (n - w(y)) \cdot \lambda_0(s) + w(y) \cdot \lambda_1(s),$$

where the corresponding eigenvector is the n-fold tensor product

$$|v_y(s)\rangle \quad := \quad |v_{y_1}(s)\rangle \otimes |v_{y_2}(s)\rangle \otimes \cdots \otimes |v_{y_n}(s)\rangle.$$

Because $\lambda_0(s) < \lambda_1(s)$ for all s, the ground state of $H(sT)$ is $|v_0(s), \ldots, v_0(s)\rangle$ with eigenvalue $n\lambda_0(s)$. The eigenvalues closest to this ground energy are those associated with the $w(y) = 1$ eigenvectors $|v_y(s)\rangle$, which have eigenvalue $(n-1)\lambda_0(s) + \lambda_1(s)$. Hence, the energy gap between the two smallest eigenvalues is $g(s) = \sqrt{2s^2 - 2s + 1}$, with its minimum $g_{\min} = \frac{1}{\sqrt{2}}$ at $s = \frac{1}{2}$ ($t = \frac{T}{2}$). Because this gap is independent of n, we can transform H_0 to H_w adiabatically with a constant delay factor. As a result, the ground state $|v_0^n(s)\rangle := |v_0(s)\rangle^{\otimes n}$ of the system evolves from $|\hat{0} \cdots \hat{0}\rangle$ to $|0 \cdots 0\rangle$ in time $O(1)$.

We will now discuss an important aspect of the above adiabatic evolution, which we will use in the lower bound of the next section. We saw how the initial ground state of the Hamiltonian H_0 is the uniform superposition $\frac{1}{\sqrt{2^n}} \sum_z |z\rangle$ while the final ground state of H_w is the zero string $|0^n\rangle$. Both states share the property that they have an exponentially small component in the subspace spanned by computational basis vectors labeled with strings of Hamming weight at least $(\frac{1}{2} + \varepsilon)n$. With the eigenvector decomposition of Equation 2 we can see that such an upper bound holds for $0 \leq s \leq 1$. Take for example the vector $|1^n\rangle$, which indeed has:

$$|\langle 1^n | v_0(s) \cdots v_0(s)\rangle| \leq \frac{1}{\sqrt{2^n}}, \tag{3}$$

for all s. This bound suggests that a perturbation of the Hamiltonian H_w in this subspace will only have an exponentially small effect on the evolution of the ground state. In the next section we will use this phenomenon to obtain an exponential lower bound on the time complexity of a perturbed version of the Minimum Hamming Weight Problem.

7.2 The Perturbed Hamming Weight Problem

We will now consider the minimization of a function that is variation of the Hamming weight function of the previous section:

$$f(z) \quad := \quad \begin{cases} w(z) & \text{if } w(z) \leq (\frac{1}{2} + \varepsilon)n, \\ p(z) & \text{if } w(z) > (\frac{1}{2} + \varepsilon)n, \end{cases} \tag{4}$$

with $\varepsilon > 0$ and $p(z)$ a decreasing function that achieves the global minimum $f(z) = p(z) = -1$ in the $w(z) > (\frac{1}{2} + \varepsilon)n$ region. Our main result will be the proof that minimum gap of the corresponding adiabatic evolution $H_f(t)$ is exponentially small, and hence that the adiabatic minimization of f requires a delay factor that is exponential in the input size n.

For clarity of exposition, we will focus on the special case where

$$f(z) \quad := \quad \begin{cases} w(z) & \text{if } z \neq 1 \cdots 1, \\ -1 & \text{if } z = 1 \cdots 1. \end{cases} \tag{5}$$

The proof contains all the ingredients required for the general result mentioned above.

The fact that this problem is a perturbed version of the Minimum Hamming Weight Problem is best expressed by

$$H_f(t) \quad := \quad H_w(t) - \frac{t}{T}(n+1)|1^n\rangle\langle 1^n|.$$

We will analyze the time-dependent eigenvalues of H_f by comparing them to those of H_w. In the previous section, we were able to diagonalize the H_w matrix by the unitary transformation $V(s)$ that maps the bit string $|y_1\rangle \otimes \cdots \otimes |y_n\rangle$ to the tensor product $|v_{y_1}(s)\rangle \otimes \cdots \otimes |v_{y_n}(s)\rangle$. Hence, using $s := \frac{t}{T}$, we have that $V^\dagger(s) \cdot H_w(t) \cdot V(s)$ is a diagonal matrix with spectrum $\{\lambda_y(s) | y \in \{0,1\}^n\}$. By looking at H_f in the eigenbasis of H_w we get the following matrix A, where we surpress some of the parameters t and s for ease of notation:

$$A \quad := \quad V^\dagger \cdot H_w \cdot V - s(n+1)V^\dagger|1^n\rangle\langle 1^n|V.$$

Note first that for $t = 0$ and $t = T$, A is a diagonal matrix. For intermediate values of t, A will have off-diagonal entries caused by the perturbation $-s(n+1)|1^n\rangle\langle 1^n|$ in the Hamiltonian H_f. At $t = 0$ the minimum eigenvalue is zero, which is indicated by the $A_{1,1} = 0$ in the top-left corner of the Hamiltonian. At $t = T$, the minimal eigenvalue has changed to -1 (for $z = 1^n$), which coincides with the bottom-right element $A_{2^n,2^n} = -1$. The eigenvectors of these values are $|v_0^n(0)\rangle$ and $|v_1^n(1)\rangle$, respectively. Intuitively, one expects the critical moment in the time evolution of H_f to occur when the ground state has to change from $|v_0^n\rangle$ to $|v_1^n\rangle$. This is indeed the case as we will see next.

To prove our claim we will introduce another matrix B that equals the matrix A with its entries $A_{2,1}, \ldots, A_{2^n,1}$ and $A_{1,2}, \ldots, A_{1,2^n}$ erased:

$$B \quad := \quad \begin{pmatrix} A_{1,1} & 0 & \cdots & 0 \\ \hline 0 & A_{2,2} & \cdots & A_{2,2^n} \\ \vdots & \vdots & & \vdots \\ 0 & A_{2^n,2} & \cdots & A_{2^n,2^n} \end{pmatrix},$$

or, equivalently,

$$B \quad := \quad A - A|0^n\rangle\langle 0^n| - |0^n\rangle\langle 0^n|A + 2A_{1,1}|0^n\rangle\langle 0^n|.$$

By construction, the state $|v_0^n(s)\rangle$ will be an eigenstate of B for every s with $A_{1,1}$ as its eigenvalue. At $t = 0$ the minimum eigenvalue of B coincides with this $A_{1,1} = 0$ entry; while at the final $t = T$ the minimum eigenvalue (with value -1) is 'located' in the $(2^n - 1) \times (2^n - 1)$ sub-matrix (corresponding to the subspace orthogonal to $|v_0^n(s)\rangle$). Because B transforms continuously between these two extremes, it follows that there is a critical moment s_c for which the minimum eigenvalue in this subspace and the eigenvalue $A_{1,1}$ are identical. In short, at s_c the matrix B has a 'zero gap' between its two minimum eigenvalues.

It can also be shown by the definitions of A and V, the fact that $V^\dagger H_w V$ is diagonal, and the lower bound of Equation 3 that:

$$\begin{aligned} \|A - B\|_2 \quad &= \quad \sqrt{2} \cdot \|A|0^n\rangle - \langle 0^n|A|0^n\rangle|0^n\rangle\|_2 \\ &= \quad s\sqrt{2}(n+1) \cdot |\langle 1^n|V(s)|0^n\rangle| \\ &\leq \quad \frac{s(n+1)}{\sqrt{2^{n-1}}}. \end{aligned}$$

The *optimal matching distance* between A and B expresses how close the spectra $\{\lambda_1, \ldots, \lambda_{2^n}\}$ and $\{\mu_1, \ldots, \mu_{2^n}\}$ of A and B are, and is formally defined by

$$d(A, B) \quad := \quad \min_\pi \max_{1 \leq j \leq 2^n} |\lambda_j - \mu_{\pi(j)}|,$$

with the minimization over all permutations $\pi \in S_{2^n}$. It is a known result in matrix analysis that for Hermitian matrices A and B this distance is upper bounded by $\|A - B\|_2$ (see Section VI.3 in [3]).

We thus reach the conclusion that for all values of s, the gap $g(s)$ of A (and hence of $H_f(s)$) will never be bigger than the gap of B plus twice the distance $\|A - B\|_2$. At the critical moment s_c, when the two minimal eigenvalues of B are identical, this implies for the gap of A the upper bound $g(s_c) \leq s_c(n+1)/\sqrt{2^{n-3}}$, and hence also for the Hamiltonian H_f: $g_{\min} \in O(\frac{n}{\sqrt{2^n}})$. Applying the requirement $T \gg g_{\min}^{-2}$ thus yields the lower bound $\Omega(\frac{2^n}{n^2})$ for the delay factor T.

7.3 Generalization

It is not difficult to see that the above lower bound method applies to the larger class of functions mentioned in Equation 4. The critical property of f is that it only deviates from the Hamming weight function $w(z)$ for those strings z that have an exponential small inner-product with the $H(s)$ ground state $|v_0(s) \cdots v_0(s)\rangle$ for all s (the property of Equation 3).

As long as the perturbation $p : \{0, 1\}^n \to \mathbb{R}$ in Equation 4 is polynomial in n, we have an inequality similar to Equation 3:

$$\|(H_f - H_w)|v_0(s) \cdots v_0(s)\rangle\|_2 \quad \in \quad 2^{-\Omega(n)}. \tag{6}$$

Hence, if the perturbation p is such that the minimum of f is not $f(0^n)$, then the adiabatic algorithm requires a delay $T \gg g_{\min}^{-2}$ that is exponential in the input size of the problem. i.e. $T \in 2^{\Omega(n)}$.

8 Conclusions

Adiabatic quantum computation is a novel paradigm for the design of quantum algorithms — it is truly quantum in the sense that it can be used to speed up searching by a quadratic factor over any classical algorithm. On the question of whether this new paradigm may be used to efficiently solve NP-complete problems on a quantum computer — we showed that the usual query complexity arguments cannot be used to rule out a polynomial time solution. On the other hand, we argue that the adiabatic approach may be thought of as a kind of 'quantum local search'. We designed a family of minimization problems that is hard for such local search heuristics, and established an exponential lower bound for the adiabatic algorithm for these problems. This provides insights into the limitations of this approach. In an upcoming paper [5], we generalize these techniques to show a similar exponential slowdown for 3SAT. It remains an open question whether adiabatic quantum computation can establish an exponential speed-up over traditional computing or if there exists a classical algorithm that can simulate the quantum adiabatic process efficiently.

9 Acknowledgements

References

[1] S. Arora, R. Impagliazzo, and U. Vazirani, "On the Role of the Cook-Levin Theorem in Complexity Theory", manuscript (1994)

[2] C. Bennett, E. Bernstein, G. Brassard, and U. Vazirani, "Strengths and weaknesses of quantum computing", *SIAM Journal on Computing,* Volume 26, No. 5, pages 1510–1523 (1997); quant-ph archive, report no. 9701001

[3] R. Bhatia, *Matrix Analysis,* Graduate Texts in Mathematics, Volume 169, Springer-Verlag (1997)

[4] M. Born and V. Fock, "Beweis des Adiabatensatzes", *Zeitschrift für Physik,* Volume 51, pp. 165–180 (1928)

[5] W. van Dam and U. Vazirani, "On the Power of Adiabatic Quantum Computation", in preparation

[6] E. Farhi, J. Goldstone, S. Gutmann, and M. Sipser, "Quantum Computation by Adiabatic Evolution", quant-ph report no. 0001106 (2000)

[7] E. Farhi, J. Goldstone, S. Gutmann, J. Lapan, A. Lundgren, and D. Preda, "A Quantum Adiabatic Evolution Algorithm Applied to Random Instances of an NP-Complete Problem", *Science,* Volume 292, April, pp. 472–476 (2001)

[8] E. Farhi, J. Goldstone, S. Gutmann "A Numerical Study of the Performance of a Quantum Adiabatic Evolution Algorithm for Satisfiability", quant-ph report no. 0007071 (2000)

[9] E. Farhi and S. Gutmann, "Analog analogue of digital quantum computation", *Physical Review A,* Volume 57, Number 4, pp. 2403–2406 (1998); quant-ph report no. 9612026

[10] L. Grover, "A fast quantum mechanical algorithm for database search", *Proceedings of the 28th Annual ACM Symposium on the Theory of Computing,* pp. 212–219 (1996); quant-ph report no. 9605043

[11] T. Hogg, "Quantum Search Heuristics", *Physical Review A,* Volume 61, Issue 5, pp. 052311 (2000)

[12] A. Messiah, *Quantum Mechanics,* John Wiley & Sons (1958)

[13] P. Shor. "Algorithms for quantum computation: Discrete logarithms and factoring", *SIAM Journal on Computing,* Volume 26, No. 5, pp. 1484–1509 (1997); quant-ph report no. 9508027

[14] U. Vazirani. "On the Power of Quantum computation", *Philosophical Transactions of the Royal Society of London A,* Volume 356, Number 1743, pp. 1759–1768 (1998)

A Proof of the Query Complexity Result

Theorem 2 *Let $b \in \{0,1\}^n$ and let I be the subset of $\{1, \ldots, n\}$ such that $b_i = 1 \iff i \in I$. Then,*

$$F(b) \quad = \quad F(0^n) + \sum_{i \in I} Y_i + \sum_{i < j \in I} Y_{ij} + \sum_{i < j < k \in I} Y_{ijk}.$$

Furthermore,

$$
\begin{aligned}
Y_i &= F(e^i) - F(0^n) \\
Y_{ij} &= F(e^{ij}) - F(e^i) - F(e^j) + F(0^n) \\
Y_{ijk} &= F(e^{ijk}) - F(e^{ij}) - F(e^{ik}) - F(e^{jk}) + F(e^i) + F(e^j) + F(e^k) - F(0^n).
\end{aligned}
$$

Proof: We count the total number of unsatisfied clauses by analyzing each type of clause.

Firstly, the only clauses of the form $\overline{X}\,\overline{X}\,\overline{X}$ that will not be satisfied are those that have all three variables with indices in I. This gives us

$$\sum_{i < j < k \in I} |\overline{X_i X_j X_k}|$$

unsatisfied clauses of the form $\overline{X}\,\overline{X}\,\overline{X}$. Note that if there are less than 3 ones in b then any of the summations over $i, j, k \in I$ satisfying $i < j < k$ will be empty and thus sum to 0.

Secondly, the only clauses of the form $\overline{X}\,\overline{X}X$ that will not be satisfied are those that have both of the negated variables with indices in I and the positive variable with index not in I. This gives us

$$\sum_{i < j \in I} |\overline{X_i X_j}X| - \sum_{i < j < k \in I} \left(|\overline{X_i X_j}X_k| + |\overline{X_i X_k}X_j| + |\overline{X_j X_k}X_i| \right)$$

unsatisfied clauses of the form $\overline{X}\,\overline{X}X$.

Thirdly, the only clauses of the form $\overline{X}XX$ that will not be satisfied will be those that have the negated variable with index in I and the positive variables with indices not in I. This gives us

$$\sum_{i \in I} |\overline{X_i}XX| - \sum_{i < j \in I} \left(|\overline{X_i}X_j X| + |\overline{X_j}X_i X| \right) + \sum_{i < j < k \in I} \left(|\overline{X_i}X_j X_k| + |\overline{X_j}X_i X_k| + |\overline{X_k}X_i X_j| \right)$$

unsatisfied clauses of the form $\overline{X}XX$.

The only clauses of the form XXX that will not be satisfied are those that contain no variable with index in I. This gives us

$$|XXX| - \sum_{i \in I} |X_i XX| + \sum_{i < j \in I} |X_i X_j X| - \sum_{i < j < k \in I} |X_i X_j X_k|$$

unsatisfied clauses of the form XXX.

Similarly, we have

$$\sum_{i < j \in I} |\overline{X_i X_j}|$$

unsatisfied clauses of the form $\overline{X}\,\overline{X}$,

$$\sum_{i \in I} |\overline{X_i}X| - \sum_{i < j \in I} \left(|\overline{X_i}X_j| + |\overline{X_j}X_i| \right)$$

unsatisfied clauses of the form $\overline{X}X$,

$$|XX| - \sum_{i \in I} |X_i X| + \sum_{i < j \in I} |X_i X_j|$$

unsatisfied clauses of the form XX,

$$\sum_{i\in I} |\overline{X_i}|$$

unsatisfied clauses of the form $\overline{X}$, and

$$|X| - \sum_{i\in I} |X_i|$$

unsatisfied clauses of the form X.

These account for all the unsatisfied clauses. Summing these quantities while rearranging terms according to the number of variables in the summations, gives us the first part of the theorem:

$$
\begin{aligned}
F(b) &= |XXX| + |XX| + |X| + \sum_{i\in I} Y_i + \sum_{i<j\in I} Y_{ij} + \sum_{i<j<k\in I} Y_{ijk} \\
&= F(0^n) + \sum_{i\in I} Y_i + \sum_{i<j\in I} Y_{ij} + \sum_{i<j<k\in I} Y_{ijk}.
\end{aligned}
$$

Notice that for $F(e^i)$ any of the summations with more than one variable will be empty, and we get

$$F(e^i) = F(0^n) + Y_i.$$

Similarly, for $F(e^{ij})$ any of the summations with three variables will be empty, and we are left with

$$F(e^{ij}) = F(0^n) + Y_i + Y_j + Y_{ij}.$$

Lastly, for $F(e^{ijk})$ we get

$$F(e^{ijk}) = F(0^n) + Y_i + Y_j + Y_k + Y_{ij} + Y_{ik} + Y_{jk} + Y_{ijk}.$$

From these equations follow the second part of the theorem. ■