Calculating the Thermal Rate Constant with Exponential Speed-Up on a Quantum Computer

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It is shown how to formulate the ubiquitous quantum chemistry problem of calculating the thermal rate constant on a quantum computer. The resulting exact algorithm scales exponentially faster with the dimensionality of the system than all known “classical” algorithms for this problem.

I. INTRODUCTION

It is well known that an exact calculation of the thermal rate constant is a problem that scales exponentially with the number of degrees of freedom. But what if we had a quantum computer (QC) at our disposal? As we will show here, then the calculation can be speeded up exponentially. Exponential speed ups on QCs have been demonstrated in a number of problems, the most famous of which are Shor’s algorithm for the factoring problem and Grover’s algorithm for database search. In the context of physics problems, exponential speed up has been demonstrated mostly in the context of simulation of the many-body Schrödinger equation. Other physics applications have also been proposed, such as studying quantum chaos and Ising spin glasses. At the experimental level QCs are still in a stage of infancy, although very impressive first steps towards implementation have been taken using ions in ion traps, atoms in high-finesse microwave cavities and molecular spins in NMR. In particular, Chuang et al., using a chloroform NMR-QC, recently implemented for the first time a quantum algorithm (Grover’s) which outperforms any classical algorithm designed to solve the same task. Another algorithm for which QCs offer an exponential speed up compared to classical computers, known as “Deutsch’s problem,” has also been implemented on a chloroform NMR-QC and by Jones et al. on a cytosine NMR-QC. These impressive achievements signal quite clearly, albeit for very simple applications at this point, that QCs may perhaps sooner than expected play an important role in simulations. For a comprehensive introduction to quantum computation, we refer the interested reader to a number of recent reviews.

The extensive body of work in quantum computation has, however, to date not addressed a computational problem of direct relevance to the quantum chemistry community (apart, of course, from the general simulation of the Schrödinger equation). In this work we show how to formulate the ubiquitous quantum chemistry problem of calculating the thermal rate constant on a QC, and by doing so, how the calculation can be speeded up exponentially with the number of degrees of freedom. The rate constant is the single most important number characterizing chemical reactions, and thus great efforts have been invested in designing efficient and exact “classical” computational ways to obtain it. Important progress along this line has been made by Miller et al., Light et al., and Manthe et al. based on the efficient evaluation of the flux correlation function. Approximate methods have also been developed for obtaining the rate constants. For example, the popular mixed quantum-classical model, whereby one integrates the time-dependent Schrödinger equation for (a few) degrees of freedom that are treated quantum mechanically, simultaneously with the classical equations of motion for the (many) degrees of freedom that are treated by classical mechanics; the semiclassical initial value representation (SC-IVR) that has had a rebirth of interest as a way for including quantum effects in molecular dynamics simulations; and the (further) linearizing approximation to the SC-IVR which leads to a much simpler form for the rate expression. While these methods enjoy favorable computational scaling properties, they are inherently approximate and thus not in the context of the present paper.

In spite of these significant advances, exact classical algorithms can at most achieve a polynomial speed up in a problem that inherently scales exponentially. Indeed, when classical algorithms are described as $O(N^3)$ instead of $O(N^2)$, it should be remembered that $N$ is itself exponentially large! The anticipated advance in quantum computation would therefore have revolutionary consequences for quantum chemistry, rendering “classical” simulation methods essentially obsolete. Of course, QCs are still at best many years away from reaching the point of replacing classical computers. Nevertheless, it is of considerable interest to exhibit an explicit QC algorithm for a problem as central as the computation of the rate constant, and this is the task we undertake here. The paper is organized as follows: In Sec. II we introduce some pertinent concepts of quantum computation. Then, in Sec. III we briefly rederive the exact quantum expression for the rate constant. The next sections are the heart of the paper, where the QC algorithm for calculation of the rate constant is described in detail. Sec. IV concludes.
II. A BRIEF INTRODUCTION TO QUANTUM COMPUTATION

Let us denote all the degrees of freedom of the problem at hand by the collective variable $\vec{q} \equiv (q_1, ..., q_M)$, and let us assume for simplicity that after discretization the number of points per degree of freedom is $2^l$. Then the Hilbert space for a given Hamiltonian $H(q, p)$ is of dimension

$$N = 2^l, \quad \nu = lM$$

whence the exponential scaling. To set up the problem on a QC one introduces a “register” of $\nu$ “qubits” (two-level systems), which can be in a superposition state

$$|\Phi\rangle = \sum_{i_1, i_2, ..., i\nu} a_{i_1} a_{i_2} ... a_{i\nu} |i_1 i_2 ... i\nu\rangle$$

Initially the quantum register $l$ qubits corresponds to one of the degrees of freedom $q_j$. Initially the quantum register $|\Phi\rangle$ is in a direct product state with all qubits in the $|0\rangle$ state:

$$|\Phi\rangle = \otimes_{i=1}^{\nu} |0_i\rangle.$$  

(2)

Allowed operations on the register are all the unitary transformations (corresponding to propagation of the register), and all measurements, i.e., projections onto subspaces of the full register Hilbert space $\mathcal{H}$. However, by convention the unitary transformations should be explicitly given in terms of operations on single and two-qubit gates. Also, allowing arbitrarily large gates would not constitute a general-purpose computer (this is similar to the situation with classical computers, where the number of distinct logical elements is a small and finite set). Simplifying an early construction by Deutsch it has been proven that the set of all single-qubit gates [the group $U(2)$] with in addition the “controlled-not” (CNOT)

$$|\alpha, \beta\rangle \rightarrow |\alpha, \beta \rangle$$

form a universal set of gates: all unitary transformations, of arbitrary size, can be constructed using a polynomially large set of the 1- and 2-qubit gates. Therefore the restriction to this set of gates is sufficient to simulate any computable function. This, of course, includes (and potentially exceeds) anything that is computable on a classical computer.

Now, let us see how to set up a superposition state on the quantum register corresponding to all possible initial classical positions. In the basis $|0\rangle = (1)$ and $|1\rangle = (0)$, one applies the one-qubit unitary “Hadamard transform”

$$R = \frac{1}{\sqrt{2}} \begin{pmatrix} 1 & 1 \\ 1 & -1 \end{pmatrix},$$

(4)

which is a $\pi/2$ rotation on the Bloch sphere of each qubit. Thus:

$$|\Phi\rangle \rightarrow |\Phi'\rangle = \otimes_{i=1}^{\nu} R|0_i\rangle = \otimes_{i=1}^{\nu} \frac{|0_i\rangle + |1_i\rangle}{\sqrt{2}} = \frac{1}{\sqrt{N}} \sum_{j=0}^{N-1} |j\rangle,$$

(5)

where $j$ is the decimal representation of the register state with the corresponding binary value, so $|j\rangle$ is a convenient shorthand notation for a tensor product of $\nu$ single-particle states. Eq.(3) represents the desired superposition over all initial positions. For example, let $l = 2$ (so the number of grid points per degree of freedom is 4) and $M = 2$ (e.g., $q_1 = x$ and $q_2 = y$ in a 2-dimensional problem involving a linear triatomic vibrating molecule) so $N = 16$; how is $|\phi\rangle = (2\Delta x, 3\Delta y)$ represented in the register? In binary, $|\phi\rangle = (|11\rangle \otimes |00\rangle) \otimes (|11\rangle \otimes |11\rangle)$, so that $j = 11$, i.e., $|\phi\rangle = |1\rangle \otimes |0\rangle \otimes |1\rangle \otimes |1\rangle$, corresponds to $x = 2\Delta x$ and $y = 3\Delta y$. In this way the quantum register supports a superposition over all discretized values of the degrees of freedom $\vec{q}$. By linearity, unitary evolution of the register amounts to parallel propagation on all of the exponentially many grid points. We will find it convenient to work from now on with $j$ as the collective degrees of freedom variable, instead of $\vec{q}$. The transformation between the two is straightforward. Note further that the register states $|j\rangle$ are position eigenstates.

Suppose now that the initial wavefunction $|\Psi(0)\rangle$ in our scattering problem has amplitude $a_j$ to be at position $j$. As explained below, it is in fact not essential to utilize this initial condition, and the quantum state at $t = 0$ is sufficient to simulate any computable function. This, of course, includes (and potentially exceeds) anything that is computable on a classical computer.

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Suppose now that the initial wavefunction $|\Psi(0)\rangle$ in our scattering problem has amplitude $a_j$ to be at position $j$. As explained below, it is in fact not essential to utilize this initial condition, and the equal superposition of all possible position eigenstates will be sufficient in most cases. Nevertheless, as shown by Zalka (and will not be repeated here), it is possible to initialize the register to the state

$$|\Phi\rangle \rightarrow |\Phi''\rangle = \sum_{j=0}^{N-1} a_j |j\rangle,$$

(6)

by means of a suitable unitary transformation. This then represents $|\Psi(0)\rangle$ on the QC. The dynamics in the scattering problem is determined by the unitary propagator $U = e^{-iHt/\hbar}$ : $|\Psi(t)\rangle = U|\Psi(0)\rangle$. The crucial advantage offered by a QC is that, as will be shown below, it is possible to efficiently implement this propagator on the quantum register, so that:

$$|\Phi\rangle \rightarrow |\Phi''\rangle = U|\Phi''\rangle.$$  

(7)

In this way we have set up a 1-1 correspondence between the QC ($|\Phi\rangle$) and the dynamics of the problem of interest ($|\Psi\rangle$). All the relevant information can be extracted from this simulation by observing the states of the qubits.
III. THE THERMAL RATE CONSTANT VIA THE FLUX CORRELATION FUNCTION FORMALISM

Let us now turn to the scattering problem and define a flux operator

\[ \mathbf{F} = \frac{i}{\hbar} [\mathbf{H}, \mathbf{h} s(\bar{q})] , \]  

(8)

where \( \mathbf{h} \) is the Heaviside function and the condition \( s(\bar{q}) = 0 \) defines the dividing surface.

The thermal rate constant is written as the time integral of the flux-flux autocorrelation function,

\[ k(T) = \frac{1}{Q_r(T)} \int_0^\infty dt C_f(t) , \]  

(9)

where \( Q_r(T) \) is the reactant partition function per unit volume, and:

\[ C_f(t) = \text{Tr}[e^{-\beta \mathbf{H}/2} \mathbf{F} e^{-\beta \mathbf{H}/2} e^{i\mathbf{H}t/\hbar} \mathbf{F} e^{-i\mathbf{H}t/\hbar}] = \text{Tr}[\mathbf{F} e^{i\mathbf{H}t/\hbar} e^{-\beta \mathbf{H}/2} \mathbf{F} e^{-i\mathbf{H}t/\hbar} \beta \mathbf{H}/2] \]

(10)

and where we have also defined for convenience the complex “time”

\[ \tau = t - i\hbar \beta / 2 . \]  

(11)

Evaluating the trace in the energy eigen-basis \( \{ |n\rangle \} \) (with \( \mathbf{H}|n\rangle = E_n |n\rangle \)) we obtain:

\[ C_f(t) = \sum_n \langle n| \mathbf{F} e^{i\mathbf{H}\tau/\hbar} \mathbf{F} e^{-i\mathbf{H}\tau/\hbar} |n\rangle \]

\[ = \sum_{n,m} \langle n| \mathbf{F} e^{i\mathbf{H}\tau/\hbar} |m\rangle \langle m| \mathbf{F} e^{-i\mathbf{H}\tau/\hbar} |n\rangle \]

\[ = \sum_{n,m} e^{iE_m\tau/\hbar} e^{-iE_n\tau/\hbar} \langle n| \mathbf{F} |m\rangle \langle m| \mathbf{F} |n\rangle \]

\[ = \sum_{n,m} e^{-\beta(E_m + E_n)/2} e^{i(E_m - E_n)t/\hbar} \langle n| \mathbf{F} |m\rangle^2 . \]  

(12)

Using the commutator form for \( \mathbf{F} \), Eq. (8), we find:

\[ \langle n| \mathbf{F} |m\rangle = \frac{i}{\hbar} \langle n| \mathbf{H} \mathbf{h} s(\bar{q}) \mathbf{h} s(\bar{q}) \mathbf{H} |m\rangle \]

\[ = \frac{1}{\hbar} \langle E_n - E_m |n\rangle \mathbf{h} s(\bar{q}) |m\rangle . \]  

(13)

Recall from Eq. (8) that the quantum register naturally supports a superposition over position eigenstates. Accordingly, let us represent \( |n\rangle \) in the discretized position basis \( \{ |j\rangle \} \), which we from now on identify with the QC’s “computational basis” \( \{ |j\rangle \} \) (indeed, the correspondence is 1-1). Thus, let us expand the energy eigenstates as:

\[ |n\rangle = \sum_{j=0}^{N-1} a_j(n) |j\rangle . \]  

(14)

Clearly, the Heaviside function \( \mathbf{h}[s(\bar{q})] \) is diagonal in this basis, so that from Eq. (13):

\[ \langle n| \mathbf{F} |m\rangle = \frac{i}{\hbar} (E_n - E_m) \sum_{j=0}^{N-1} a_j^*(n) a_j(m) \mathbf{h}[s(j)] . \]  

(15)

Hence, finally:

\[ C_f(t) = \frac{1}{\hbar^2} \sum_{n,m \neq m} e^{-\beta(E_m + E_n)/2} e^{i(E_m - E_n)t/\hbar} (E_n - E_m)^2 \]

\[ \times \sum_{j=0}^{N-1} a_j^*(n) a_j(m) \mathbf{h}[s(j)]^2 . \]  

(16)

Our task is therefore to find an algorithm that calculates the spectrum \( \{ E_n \} \), and the position amplitudes \( \{ a_j(n) \}_{j=0}^{N-1} \) for each of the eigenstates \( |j\rangle \). With these in hand the rest of the calculation [summations in Eq. (16)] can be efficiently implemented on a classical computer.

At first sight it appears that there are two problems: (1) The spectrum may contain exponentially many energies; (2) The number of measurements needed must be exponentially large (even if the number of energy eigenstates is polynomial) since there are exponentially many position eigenstates \( |j\rangle \)!

Regarding (1), the summation should indeed extend over a polynomially large number of energies only. This, however, is reasonable if the spectrum is sufficiently degenerate and, more importantly, since the exponential decrease due to the Boltzman factors will effectively eliminate the higher end of the spectrum \( E \gtrsim k_B T + \text{(barrier height)} \) even if it is not degenerate. As for (2), the question is how to get a reasonable sample of the distribution. For example, if the distribution is badly behaved in position space, one could Fourier transform to momentum space and sample there. At any rate, this problem is identical to the one faced by any classical simulation, where one settles for a Monte Carlo sampling of the wavefunction. This is not the source of the classical bottleneck, and we will adopt the same Monte Carlo approach in the quantum simulation. Note, however, that unlike the classical case where attention has to be given to the issue of generating statistically independent Monte Carlo samples, the quantum simulation automatically generates truly independent samples by the projection postulate.

How can we efficiently calculate the eigenstates and the spectrum? Solving the time-independent Schrödinger equation (SE)

\[ \mathbf{H}|\psi\rangle = E|\psi\rangle \]  

(17)

can be done on a QC by transforming the problem to the time-dependent SE and propagating the dynamics with the unitary time-evolution operator \( \mathbf{U} = e^{-i\mathbf{H}t} \). Each energy eigenvalue and eigenstate can then be obtained via known quantum algorithms, to be detailed below, in polynomial time.
IV. GENERAL OUTLINE OF THE ALGORITHM

We now give the algorithm in general terms, to be defined more precisely in the next section.

1. Prepare a register as in Eq. (5), and attach some “ancilla” qubits to it, also in the \( |0\rangle \) state. These will serve as a quantum scratch-pad to record the results of intermediate measurements. From now on we will distinguish between the “main” and ancillary registers.

2. If a good guess for the initial wavefunction is known, initialize the register to it as in Eq. (6). Else initialize the register to an equal superposition. Since the computational basis states are position eigenstates in all likelihood they are not energy eigenstates, so will not be stationary under the SE dynamics. Thus except if the equal superposition corresponds to some undesirable position – such as very high above the barrier so that dissociation sets in immediately – it is as good a guess as any. In fact, any random (but reproducible) initial distribution will do.

3. “Propagate” the register in parallel for a time \( t \). This corresponds to a parallel evolution of all the position eigenstates. The propagation is done very much in analogy to the classical FFT method \[27\] in particular the split time propagation scheme \[28\]. Namely, the potential part is diagonal and can be implemented directly, whereas for the kinetic part it is necessary to Fourier transform to and back from momentum space.

4. Perform a “von Neumann” measurement (see Sec. \ref{sec:inputting}) on the ancillary register using the Hamiltonian (energy) as the observable. This accomplishes a double purpose:
   (a) It allows to obtain an energy \( E_n \) by measuring the ancillas.
   (b) It provides a means to sample the energy-position amplitudes \( a_j(n) \).

5. Repeat steps 1-4 many times until the distribution is converged to the desired accuracy for all relevant eigenstates. The number of required repetitions is proportional to this accuracy.

6. Calculate (classically) the sums in Eq. (16).

V. THE ALGORITHM IN DETAIL

A. Initialization

Here the register is initialized to the state \( |\Phi\rangle = \bigotimes_{i=1}^{2\nu} |0_i\rangle \), where the last \( \nu \) qubits are ancillas. The physics of this initialization step depends on the QC implementation. One conceivable way is cooling to the ground state.

B. Inputting the Initial Wavefunction

If necessary one inputs the initial wavefunction by the technique of Zalka.\[29\] Else one employs the Hadamard rotations technique to create an equal superposition over position states, as in Eq. (6). In the former case the register will be in the state:

\[
|\Phi''\rangle = \left( \sum_{j=0}^{N-1} \alpha_j |j\rangle \right) \bigotimes_{i=\nu}^{2\nu} |0_i\rangle. \tag{18}
\]

In the latter case all \( \alpha_j = 1 \).

C. Quantum Propagation Algorithm

This subsection is the heart of the algorithm; it builds on the approach of Zalka. Assume for simplicity that we have a single particle of mass \( m \) in an external potential \( V(q) \). The full Green’s function for arbitrary time \( t \) is:

\[
G(x_1, x_2; t) = \langle x_1 | e^{-iHt/\hbar} | x_2 \rangle. \tag{19}
\]

For short time steps \( \Delta t \ll 1/E \) (\( E \) is a typical energy of the system) this becomes approximately:

\[
G(x_1, x_2; \Delta t) = \kappa \exp \left[ im \frac{(x_1 - x_2)^2}{2\Delta t} - iV(x_1)\Delta t \right], \tag{20}
\]

where \( \kappa \) is a normalization factor. Applying this to the amplitudes is equivalent to acting on the basis states with the inverse transformation. Thus the position eigenstates, properly normalized, transform as:

\[
|j\rangle \mapsto U|j\rangle = \frac{1}{\sqrt{N}} \sum_{j'=0}^{N-1} \exp \left[ -i m \frac{(j - j')^2 \Delta x^2}{2\Delta t} + iV(j\Delta x)\Delta t \right] |j'\rangle. \tag{21}
\]

This is carried out in parallel on the entire superposition \[ \sum_{j=0}^{N-1} |j\rangle \]. Suppose the time-step and spatial resolution are adjusted so that:

\[
\frac{m \Delta x^2}{\Delta t} = \frac{2\pi}{N}. \tag{22}
\]

Then by expanding the exponent Eq. (21) can be written as a succession of a diagonal transformation, Fourier transform, and another diagonal transformation, all unitary:
\( U[j] = \exp\{iF_2(j)\} F(j, j') \exp\{iF_1(j')\} |j'\rangle \),

where:

\[
F_1(j) = -\pi j^2 / N \\
F_2(j) = -\pi j^2 / N + V(j \Delta x) \Delta t
\]

\( F(j, j') |j\rangle = \frac{1}{\sqrt{N}} \sum_{j'=0}^{N-1} \exp\left[2\pi i j j'/N\right] |j'\rangle. \)

Eq. (22) tells us how many qubits \( \nu = \log_2 N \) are needed for given \( \Delta x \) and \( \Delta t \):

\[
\nu = \log_2 \frac{2\pi \Delta t}{m \Delta x^2}.
\]

The special form of Eq. (23), involving diagonal transformations and a Fourier transform, is due to the structure of the Hamiltonian operator as a sum of operators diagonal in coordinate and momentum space. As mentioned above, this is very similar to the situation that arises in the classical FFT method for solving the SE.

1. Diagonal Transformations:

Consider first executing the diagonal unitary transformations \( |j\rangle \mapsto \exp\{iF(j)\} |j\rangle \), which can be done as follows, using the ancillary register [Eq. (18)], in the state \( |0\rangle = \bigotimes_{l=0}^{2\nu} |0_l\rangle \). The number \( \nu \) of qubits in this register depends on the accuracy with which \( F \) needs to be evaluated (see immediately below). Then the following steps are applied:

1. \( |j, 0\rangle \mapsto |j, F(j)\rangle \): evaluation of \( F \) and storage of the result in the ancillary register;
2. \( |j, F(j)\rangle \mapsto \exp\{iF(j)\} |j, F(j)\rangle \) : introducing the phase;
3. \( \exp\{iF(j)\} |j, F(j)\rangle \mapsto \exp\{iF(j)\} |j, 0\rangle \) : inversion of step 1 in order to clear the ancillary register.

Step 1 requires that it is possible to evaluate an arbitrary function and store the result. This is very similar to the equivalent classical problem, for which algorithms are known using just the elementary classical gates. The same can be done in the quantum case, by breaking up the evaluation into elementary arithmetic operations, for which quantum algorithms have been designed. We will not dwell on this issue here. Step 3 is just the reverse of step 1 and can therefore be implemented by running the inverse unitary transformation.

Step 2 has no classical analogue since it involves phases. It can be implemented if one knows how to do \( |x\rangle \mapsto \exp\{ix\} |x\rangle \). This can be done by simple single-qubit phase-shifts. Let \( \nu = 2k \). Using a binary expansion \( x = \sum_{l=-k}^{k-1} x_l 2^l \), we have: \( |x\rangle = |x_{-k} \rangle \otimes |x_{-k+1} \rangle \otimes \cdots \otimes |x_{-1} \rangle \), where \( x_l = 0, 1 \). In the standard basis \( |0\rangle = (1, 0) \), \( |1\rangle = (0, 1) \), consider the following unitary operation:

\[
Q = \bigotimes_{l=-k}^{k-1} \left( \begin{array}{cc} 1 & 0 \\ 0 & e^{2\pi i} \end{array} \right).
\]

The \( l \)th \( 2 \times 2 \) matrix is a unitary operation in the Hilbert space of qubit number \( l \). Thus:

\[
\left( \begin{array}{cc} 1 & 0 \\ 0 & e^{2\pi i} \end{array} \right) |x_l\rangle = e^{ix_l^2} |x_l\rangle.
\]

Therefore the full result is:

\[
Q |x\rangle = \bigotimes_{l=-k}^{k-1} e^{ix_l^2} |x_l\rangle = e^{i \sum_{l=-k}^{k-1} x_l^2} \bigotimes_{l=-k}^{k-1} |x_l\rangle = e^{ix} |x\rangle,
\]

as required.

2. The Quantum Fourier Transform

The quantum Fourier transform (QFT) algorithm has been discussed extensively, and some beautiful connections to group theory have been made. In view of its central importance in the algorithm for solving the SE (and indeed in all efficient quantum algorithm so far!), we present a brief derivation here, using the approach of Cleve et al.

The QFT was defined in Eq. (24). Using the binary-decimal notation \( j/N = j_1 j_2 \ldots j_{\nu} \) (recall that \( N = 2^\nu \)) where \( j_1 = 0, 1 \) etc., we note first that:

\[
e^{2\pi i j / 2^\nu} |j_1, j_2, \ldots, j_{\nu}\rangle = e^{2\pi i (0, j_2) j_1^0} |j_1^0, j_2^0, \ldots, j_{\nu}^0\rangle \]

\[
\otimes e^{2\pi i (0, j_{-1} j_0) j_2^0} |j_2^0, \ldots, j_{\nu}^0\rangle \otimes \cdots \otimes e^{2\pi i (0, j_1 j_{\nu-1}) j_{\nu}^0} |j_{\nu}^0\rangle,
\]

(29)

It follows that:

\[
\sum_{j'=0}^{N-1} \exp\left[2\pi i j j'/N\right] |j'\rangle = \\
\left( |0\rangle + e^{2\pi i (0, j_0)} |1\rangle \right) \otimes \left( |0\rangle + e^{2\pi i (0, j_{-1} j_0)} |1\rangle \right) \otimes \cdots \\
\otimes \left( |0\rangle + e^{2\pi i (0, j_1 j_{\nu-1})} |1\rangle \right),
\]

(30)

by expanding out the product on the right-hand-side and a term-by-term comparison. Thus the Fourier-transformed state in Eq. (27) is in fact an “untangled” direct product. This fact greatly simplifies the implementation of the QFT.

To perform the QFT, one first applies a Hadamard rotation [Eq. (4)] to \( |j_1\rangle \) (the first qubit of \( |j\rangle \)), with the result:
\[ R|j_1\rangle = (|0\rangle + (-1)^{j_1}|1\rangle) = (|0\rangle + e^{2\pi i(0,j_1)}|1\rangle), \]  
(31)

so: 
\[ |j\rangle \mapsto (|0\rangle + e^{2\pi i(0,j_1)}|1\rangle) |j_2,\ldots,j_\nu\rangle. \]

Let us now define a new single-qubit operation, similar to \( Q \) from Eq. (26):

\[ Q_l = \left( \begin{array}{cc} 1 & 0 \\ 0 & e^{2\pi i/2^l} \end{array} \right). \]  
(32)

This operation is applied on the first qubit \( |j_1\rangle \), subject to a control by a second qubit \( |j_2\rangle \) (which itself does not change): a “controlled rotation”. Namely, if \( j_1 = 0 \) one does nothing, if it is 1, one applies \( Q_l \). This can be written as the following unitary transformation in the 4-dimensional Hilbert space of the two qubits, in the standard basis \( |j_2,j_1\rangle = |00\rangle = (1,0,0,0), |01\rangle = (0,1,0,0), |10\rangle = (0,0,1,0), |11\rangle = (0,0,0,1): \n
\[ CQ_l = \left( \begin{array}{cc} 1 & 0 \\ 0 & |Q_l| \end{array} \right). \]  
(33)

After applying \( CQ_2 \) one obtains:

\[ (|0\rangle + e^{2\pi i(0,j_1,j_2)}|1\rangle). \]  
(34)

Next a “controlled-\( Q_3 \)” is applied, yielding:

\[ (|0\rangle + e^{2\pi i(0,j_1,j_2,j_3)}|1\rangle). \]  
(35)

Clearly, this process will eventually generate the desired phase in the superposition state of the first qubit [corresponding to the last qubit in Eq. (34)]:

\[ \left( \prod_{l=2}^{\nu} CQ_l \right) \left( |j_1\rangle = (|0\rangle + e^{2\pi i(0,j_1,j_2\ldots,j_\nu)}|1\rangle) \right). \]  
(36)

where the terms in the product from here onwards are applied low index first.

Now we turn to the second qubit. Again, a Hadamard rotation on it has the effect of: \( R|j_2\rangle = (|0\rangle + e^{2\pi i(0,j_2)}|1\rangle) \). This is followed by a controlled-\( Q_2 \), conditioned upon \( |j_3\rangle: \ (|0\rangle + e^{2\pi i(0,j_2)}|1\rangle) \mapsto (|0\rangle + e^{2\pi i(0,j_2,j_3)}|1\rangle). \)

After the full operation on \( |j_2\rangle \) one obtains:

\[ \left( \prod_{l=2}^{\nu-1} CQ_l \right) \left( |j_2\rangle = (|0\rangle + e^{2\pi i(0,j_2j_3\ldots,j_\nu)}|1\rangle) |j_2\rangle. \]  
(37)

which corresponds to the one before last qubit in Eq. (34).

The method to generate the entire product in Eq. (34) should now be clear; collecting all the transformations yields:

\[ |j\rangle \mapsto \prod_{p=1}^{\nu-1} \left( \prod_{l=2}^{\nu} CQ_l \right) \left( |j_1,\ldots,j_\nu\rangle = (|0\rangle + e^{2\pi i(0,j_1j_2\ldots,j_\nu)}|1\rangle) \otimes \ldots \right) \right) \otimes (|0\rangle + e^{2\pi i(0,j_\nu)}|1\rangle). \]  
(38)

Up to an unimportant bit reversal (which can easily be rectified by permuting the role of the qubits in the transformations above), this is exactly the desired result. In other words, the QFT is simply:

\[ F = \prod_{p=1}^{\nu-1} \left( \prod_{l=2}^{\nu} CQ_l \right) R \]  
(39)

This will be applied in parallel, by virtue of the superposition principle, on all position eigenstates \( |j\rangle \). Most importantly, the number of operations (single- and two-qubit) needed to implement the QFT is seen to be a mere \( \nu(\nu-1)/2 \). This is to be compared to the \( \nu^{2\nu} \) operations required classically, and as emphasized above, is the “secret” behind the quantum speedup.

D. von Neuman Measurements

Combining Eqs. (7) and (21), at this point the register is in the state

\[ |\Phi''\rangle = \sum_j \alpha_j U |j\rangle = \sum_{j'} \psi_{j'}(t) |j'\rangle \]

\[ \psi_{j'}(t) = \sum_j \alpha_j G^{-1}(j,j';t). \]  
(40)

A parallel propagation has occurred on all the position eigenstates. By measuring the qubits one by one, i.e., projecting onto a random position eigenstate \( |j'\rangle \), and repeating this process many times while collecting the statistics, one can sample the electronic density function \( |\psi_{j'}(t)|^2 \). Our goal was to find the energy-spectrum and energy-position amplitudes \( a_j(n) \), so these should be obtained from the simulation. This can be done using the so-called “von Neuman measurement” trick. We will require an additional propagation step.

A “measurement apparatus” that can be made to interact with the QC is introduced, and is assumed to be equivalent to a 1-dimensional quantum mechanical particle. That is, its Hilbert space is spanned by the basis vectors \( |x\rangle \), \( x \) real, with \( \mathbf{X}|x\rangle = x|x\rangle \). In practice this will be another ancillary quantum register, consisting of, say, \( K \) qubits. Now, let us expand the position eigenstates \( |j'\rangle \) in terms of the complete set of energy eigenstates [recall Eq. (14)]:

\[ |j'\rangle = \sum_n a_{j'}^n(n)|n\rangle. \]  
(41)
Consider next the joint evolution of an energy eigenstate \(|n\rangle\) and the apparatus state \(|x\rangle\) (\(x\) is arbitrary), under the unitary operator \(\tilde{U} = \exp(i\mathbf{H} Pt/h)\), where \([\mathbf{X}, \mathbf{P}] = i\hbar\). Here \(\mathbf{H}\) acts on the main register and \(\mathbf{X}, \mathbf{P}\) act on the apparatus, so \([\mathbf{X}, \mathbf{H}] = [\mathbf{P}, \mathbf{H}] = 0\). We will shortly discuss the implementation of \(\tilde{U}\). Consider first a formal Taylor expansion of \(\exp(\mathbf{H} Pt/h)\), which yields:

\[
\tilde{U}|n\rangle|x\rangle = \sum_{l=0}^{\infty} \frac{1}{l!} (tE_n)^l |n\rangle|x + tE_n\rangle. \tag{42}
\]

Thus \(\tilde{U}\) does not change the energy eigenstate, but has the effect of “shifting the dial \(x\)” by an amount proportional to the energy \(E_n\). The effect on the position eigenstate \(|j\rangle\) will be:

\[
\tilde{U}|j\rangle|x\rangle = \sum_{n} a_j^*(n) |n\rangle|x + tE_n\rangle, \tag{43}
\]

and the effect on the full superposition of Eq.(41) is:

\[
\tilde{U}|\Phi''\rangle|x\rangle = \sum_{j} \psi_j(t) \sum_n a_j^*(n) |n\rangle|x + tE_n\rangle
= \sum_{n} \xi_n(t)|n\rangle|x + tE_n\rangle
= \sum_{j} a_j^*(n) \psi_j(t). \tag{44}
\]

Now suppose we observe the state of the apparatus. From Eq.(41) it is clear that the apparatus has become entangled with the QC, and by performing the observation the superposition will collapse onto a particular state \(|m\rangle|x + tE_m\rangle\). This happens with probability \(|\xi_n(t)|^2\).

Recall that \(|x + tE_m\rangle\) is represented in binary by the qubits of the apparatus. Since \(t\) is a parameter of the simulation and \(x\) is known, all that remains is to measure the apparatus qubit by qubit, to obtain the energy eigenvalue \(E_m\). The accuracy with which these numbers are obtained is proportional to the number of simulation steps.

To implement \(\tilde{U}\) it is necessary to Fourier transform the \(|x\rangle\) register, just like in the classical FFT case. Specifically, let us define the Fourier transform pair:

\[
\mathcal{F}|x\rangle = |p\rangle = \frac{1}{\sqrt{2^K}} \sum_{x=0}^{2^K-1} e^{-ixp/h} |x\rangle
\]

\[
\mathcal{F}|p\rangle = |x\rangle = \frac{1}{\sqrt{2^K}} \sum_{p=0}^{2^K-1} e^{ixp/h} |p\rangle. \tag{45}
\]

Then starting from the initial apparatus state \(|x\rangle\), \(\tilde{U}\) can be implemented as follows:

\[
|n\rangle|x\rangle \xrightarrow{\mathcal{F}} |n\rangle|p\rangle
\xrightarrow{\tilde{U}} e^{iE_nPt/h} |n\rangle|p\rangle
\]

\[
\mathcal{F}^{-1}(p) |n\rangle \frac{1}{\sqrt{2^K}} \sum_{p=0}^{2^K-1} e^{i(x+E_n)t/p}|p\rangle
= |n\rangle|x + E_n t\rangle, \tag{46}
\]

in agreement with Eq.(42).

E. Extracting the Amplitudes from the Measurements

Note further that after observation of the apparatus, the state of the main register has been projected onto \(|m\rangle\), an energy eigenstate. The \(\tilde{U}\) propagation had a remarkable outcome: it transformed the information in the main register from a superposition over position eigenstates \(|j\rangle\) to one over energy eigenstates \(|n\rangle\). Unfortunately, apart from utilizing this in a subsequent evolution, this, however, does not appear to be particularly useful, for it is not clear how the energy eigenstates are enumerated in the main register after the von Neuman propagation. Nevertheless, it is possible to obtain the amplitudes \(a_j(n)\) needed to complete the calculation in Eq.(41). Note first that by Eq.(41):

\[
\sum_{n} \xi_n(t)a_j(n) = \psi_j(t). \tag{47}
\]

Now, the simulation yields, by performing the whole procedure a sufficient number of times, an estimate of the probabilities \(|\psi_j(t)|^2\) [Eq.(40)] and \(|\xi_n(t)|^2\) [Eq.(44)]. Thus to fully specify the complex numbers \(a_j(n)\), it is necessary to also know their phases, as well as those of the \(\xi_n(t)\) and \(\psi_j(t)\).

To obtain the phases, we note first that it is sufficient to know only the signs, since no generality is lost by employing a real initial wavefunction \(|\Psi(0)\rangle\). The signs can then be obtained with the help of a simple trick, which we will illustrate on a generic 2-qubit register state \(|\psi\rangle = a_0|00\rangle + a_1|01\rangle + a_2|10\rangle + a_3|11\rangle\). Given repeated preparations of this \(|\psi\rangle\), we perform the following set of measurements:

- Observation of the two qubits in \(|\psi\rangle\).
- A Hadamard transform on the first qubit, followed by observation of the two qubits.
- A Hadamard transform on the second qubit, followed by observation of the two qubits.

The first step yields an estimate of the \(|a_i\rangle\). The second one yields an estimate of \(|a_0 \pm a_1\rangle\) and \(|a_2 \pm a_3\rangle\), since under the Hadamard transform \(|\psi\rangle \mapsto \frac{1}{\sqrt{2}}[(a_0 + a_1)|00\rangle + (a_0 - a_1)|01\rangle + (a_2 + a_3)|10\rangle + (a_2 - a_3)|11\rangle}\). Similarly, the third step yields an estimate of \(|a_0 \pm a_2\rangle\) and \(|a_1 \pm a_3\rangle\), since \(|\psi\rangle \mapsto \frac{1}{\sqrt{2}}[(a_0 + a_2)|00\rangle + (a_0 - a_2)|01\rangle + (a_1 + a_3)|10\rangle + (a_1 - a_3)|11\rangle]\). Clearly, this provides sufficient information for extraction of the signs of all amplitudes.
The generalization to a $\nu$-bit register is obvious: one performs Hadamard rotations on all $\nu$ qubits. This then yields \{$$|a_0 \pm a_1|, |a_2 \pm a_3|, |a_4 \pm a_5|, \ldots$$\} (after Hadamard on first qubit); \{$$|a_0 \pm a_2|, |a_1 \pm a_3|, |a_4 \pm a_6|, \ldots$$\} (after Hadamard on second qubit); etc. After each Hadamard rotation there are $2^n$ coefficients to be estimated. This exponential “Monte-Carlo scaling” is the same as the one we encountered before and is not considered a slow-down for the reasons detailed above. The additional computational cost is in the Hadamard rotations, $\nu$ of which must be performed. This does therefore not affect the efficiency of the algorithm. At the end of the process, if the whole phase space has been sampled, one is left with $\nu 2^n$ absolute values equations, which contain sufficient information to solve for the signs of all the amplitudes. In practice one will of course sample only a small (polynomial) portion of the phase space, and care must then be taken to obtain sufficient equations of the type above to uniquely determine the signs of the amplitudes of interest.

F. Repetition

The steps outlined above generate the energies $$\{E_n\}$$ and estimates of amplitudes $$\{a_j(n)\}$$ needed to perform the sum in Eq. (15). The whole process must now be repeated many times, on the order of the required accuracy, in order to complete the calculation. Due to the speed up in the implementation of the propagation step, the algorithm performs exponentially faster than any exact classical algorithm designed to solve the same task.

VI. DISCUSSION AND CONCLUSIONS

QCs are still far from being a panacea, and doubts have been raised whether they will ever replace ordinary, classical computers. Such worries are invariably based on the immense difficulties associated with maintaining phase coherence throughout the computation, i.e., the “decoherence problem.” However, a remarkable theory of quantum error correction codes has recently been constructed in which a “logical qubit” is encoded in the larger Hilbert space of several physical qubits. It has been shown that as long as the error rate is sufficiently small, it is possible to perform fault-tolerant quantum computation, i.e., the computation can be stabilized and be made fully robust to errors. These advances greatly enhance the prospects of the eventual construction of useful QCs, beyond the current highly rudimentary prototypes. Building on these hopes, we have presented here an algorithm for calculating the thermal rate constant on a QC. The algorithm involves an initialization step of the QC into an equal superposition of position eigenstates; a propagation using an adaptation to QCs of the well-known FFT technique; and finally, a sequence of measurements yielding the energy spectrum and amplitudes. Under reasonable assumptions about the distribution of energy eigenvalues the algorithm runs in polynomial time. The algorithm thus outperforms any exact classical simulation, which is bound to be exponential. This clearly demonstrates the potential utility of QCs in future applications to quantum chemistry problems.

Our approach was somewhat of a “brute force” one, in that we did not attempt to optimize the algorithm using such fruitful concepts as “direct and correct” low-rank expressions for the rate constant. Such optimizations, while ineffectual in altering the essential exponential speed up achieved by use of a QC, may still be important in practice, especially in the early stages of the application on a small-scale QC of an algorithm such as described here. Further work is hence desirable to optimize the algorithm.

Finally, it would be interesting to check the effect of noise and other types of errors affecting the evolution of the QC on the present algorithm. It has been shown, e.g., in the case of the ion trap QC, that factoring becomes impossible once random phase fluctuations in the laser pulses exceed a certain threshold. We intend to study similar noise-related issues using numerical simulations in the context of the present algorithm.

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