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# Quantum computing methods for electronic states of the water molecule

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#### ABSTRACT

We compare recently proposed methods to compute the electronic state energies of the water molecule on a quantum computer. The methods include the phase estimation algorithm based on Trotter decomposition, the phase estimation algorithm based on the direct implementation of the Hamiltonian, direct measurement based on the implementation of the Hamiltonian and a specific variational quantum eigensolver, Pairwise VQE. After deriving the Hamiltonian using STO-3G basis, we first explain how each method works and then compare the simulation results in terms of gate complexity and the number of measurements for the ground state of the water molecule with different O–H bond lengths. Moreover, we present the analytical analyses of the error and the gate-complexity for each method. While the required number of qubits for each method is almost the same, the number of gates and the error vary a lot. In conclusion, among methods based on the phase estimation algorithm, the second-order direct method provides the most efficient circuit implementations in terms of the gate complexity. Moreover, Pairwise VQE serves the most practical method for near-term applications on the current available quantum computers. Finally the possibility of extending the calculation to excited states and resonances is discussed.



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# 1. Introduction

The problem at the heart of computational chemistry is electronic structure calculation. This problem concerns calculating the properties of the stationary state describing many electrons interacting with some external potential and between each other via Coulomb repulsion. The ability to efficiently solve these problems for the cases of many body systems can have huge effects in pharmaceutical development, materials engineering, and all areas of chemistry. Quantum computing proposes the possibility to efficiently solve this problem for molecules with many more electrons than what can currently be simulated by classical computers [1]. The ability to calculate properties of large quantum systems using precise control of some other quantum system was first proposed by Feynman [2]. He pointed out that if you have enough control over the states of some quantum system, you can create an analogy to some other quantum system. Using the example of spin in a lattice imitating many properties of bosons in quantum field theory, he conjectured that if you have enough individual quantum systems you could simulate any arbitrary quantum mechanical system. Simulation of the electronic structure Hamiltonian works very similar to this. Using the Jordan-Wigner or Bravyi-Kitaev transformation [3–5] you can map an electronic structure

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Hamiltonian to a spin-type Hamiltonian which preserves energy eigenvalues [6]. Evolution under this spin-type Hamiltonian,  $e^{-iHt}$ , can then be approximately simulated on quantum computers.

Quantum simulation provides a new and efficient way to calculate eigenenergies of a given molecule. Classically the problem would have a computational cost which grows exponentially with the system size, n, the number of orbital basis functions [7]. However, based on the phase estimation algorithm [8,9], the molecular ground state energies can be calculated with gate depth O(poly(n)) [10–12]. The quantum circuit for the Hamiltonian is generally approximated through a Trotter-Suzuki decomposition. It is shown that the Hamiltonian dynamics can also be simulated through a truncated Taylor series [13]. This method is generalised as quantum signal processing [14]. Babbush et al. [15] further shows that it is possible to reduce the gate depth of the circuit to O(n) by using plane wave orbitals. Recently, a direct circuit implementation of the Hamiltonian within the phase estimation (Direct-PEA) is presented by authors of paper [16–18]: the circuit designs are provided to the time evolution operator by using the truncated series such as  $U = I - iH/\kappa$  and  $U = tH + i(I - t^2H^2/2)$ , in which  $\kappa$ and t are parameters to restrict truncation error. These unitary operators are much simpler to implement than those of a Trotter decomposition, and can be also used to calculate ground state energies of molecular Hamiltonians. Another approach called variational quantum eigensolver (VQE) has been introduced by Aspuru-Guzik and coworkers [19,20]: This method combines classical and quantum algorithms together and significantly reduces the gate complexity at the cost of a large amount of measurements. It has also been applied on real-world quantum computers to solve ground state energies of molecules such as: H<sub>2</sub>, LiH and BeH<sub>2</sub> [21,22].

This paper explores all these above mentioned methods for calculating the ground state energies of the water molecule and presents a comparison study, in terms of both the accuracy and the gate complexity dependent on error. The next section explains the method by which the electronic Hamiltonian for water is calculated and the method by which to reduce the number of qubits required to simulate the transformed spin-type Hamiltonian. Then, Section 3 discusses five methods of electronic structure simulation on quantum computers: the phase estimation using first-order Trotter-Suzuki decomposed propagator (Trotter PEA), two direct implementations of the spin-type Hamiltonian (Direct PEA), a direct measurement and a specific variational quantum eigensolver method(Pairwise VQE). Section4 shows results for these methods with comparison to the exact energy from direct diagonalisation of the spin-type Hamiltonian. It also

gives qubit requirement and gate complexity for different methods asymptotically. Spin-type Hamiltonian for  $H_2O$  at equilibrium bond length is derived in Appendix 1. Details of both error and complexity analyses are given in Appendices 2 and 3.

#### 2. Hamiltonian derivation

In this section we provide details for calculating the spintype Hamiltonian describing electronic structures of the water molecule using STO-3G basis set that will be used in later methods. This derivation can be generalised to an arbitrary molecular Hamiltonian.

To obtain the Hamiltonian of the water molecule, we start by considering the 1s orbital of each hydrogen atom along with the 1s, 2s,  $2p_x$ ,  $2p_y$ ,  $2p_z$  orbitals for the oxygen atom. This leads to a total of 14 molecular orbitals considering spin. To make our simulations more efficient, the number of qubits is reduced by considering orbital energies and exploiting the symmetry of the system [22].

It can be initially assumed that the two molecular orbitals of largest energies are unoccupied. Consequently, the calculation of the Hamiltonian of the water molecule then only requires the consideration of 12 spinorbitals. After second quantisation, the Hamiltonian can be expressed as [23]:

$$H = \sum_{i,j=1}^{12} h_{ij} a_i^{\dagger} a_j + \frac{1}{2} \sum_{i,j,k,l=1}^{12} h_{ijkl} a_i^{\dagger} a_j^{\dagger} a_k a_l.$$
(1)

Here  $a_i^{\dagger}$  and  $a_i$  are fermionic creation and annihilation operators, and  $h_{i,j}$  and  $h_{i,j,k,l}$  are one-body and twobody interaction coefficients. In this work the molecular orbitals are calculated from the Hartree-Fock method and represented by the STO-3G basis functions. The numerical integration obtaining the one and two electron integrals for molecular water is performed by the PyQuante package [24]. The expressions for these integrations are:

$$h_{ij} = \int d\vec{r}_1 \chi_i^*(\vec{r}_1) \left( -\frac{1}{2} \nabla_1^2 - \sum_{\sigma} \frac{Z_{\sigma}}{|\vec{r}_1 - \vec{R}_{\sigma}|} \right) \chi_j(\vec{r}_1),$$
(2)

$$h_{ijkl} = \int d\vec{r}_1 d\vec{r}_2 \chi_i^*(\vec{r}_1) \chi_j^*(\vec{r}_2) \frac{1}{r_{12}} \chi_k(\vec{r}_2) \chi_l(\vec{r}_1).$$
(3)

Here we have defined  $\chi_i(\vec{r})$  as the *i*th spin-orbital, which is calculated from a spatial orbital obtained by the Hartree-Fock method and the electron spin states.  $Z_{\sigma}$  is the  $\sigma$  th nuclear charge,  $\vec{r_i}$  is the position of electron *i*,  $r_{12}$ is the distance between the two points  $r_1$  and  $r_2$ , and  $\vec{R}_{\sigma}$ is the position of  $\sigma$  th nucleus. We have ordered our spin-orbitals from 1 to 12 as follows:  $\{1 \uparrow, 2 \uparrow \dots, 6 \uparrow, 1 \downarrow, 2 \downarrow, \dots 6 \downarrow\}$ , with first spinup orbitals ordered from lowest to highest energy and continuing into spin-down orbitals ordered from lowest to highest energy. Now introduce an ad hoc set F = $\{1, 2, 7, 8\}$  corresponding to the 4 lowest energy spin orbitals  $\{1 \uparrow, 2 \uparrow, 1 \downarrow, 2 \downarrow\}$ . For the H<sub>2</sub>O ground state, it can be assumed the spin orbitals in the set *F* will be filled with electrons. The following one-body single electron interaction operators then become:

$$a_{1}^{\dagger}a_{1} = 1, \quad a_{2}^{\dagger}a_{2} = 1, \quad a_{7}^{\dagger}a_{7} = 1, \quad a_{8}^{\dagger}a_{8} = 1,$$
  
 $a_{i}^{\dagger}a_{j} = 0, \quad \text{if } i \neq j \quad \text{and} \quad i \in F \text{ or } j \in F.$  (4)

This assumption also allows us to simplify the twoelectron interaction terms under certain conditions:

$$a_{i}^{\dagger}a_{j}^{\dagger}a_{k}a_{l} = \begin{cases} a_{j}^{\dagger}a_{k}, & i = l, i \in F, \{j, k\} \notin F, \\ a_{i}^{\dagger}a_{l}, & j = k, j \in F, \{i, l\} \notin F, \\ -a_{j}^{\dagger}a_{l}, & i = k, i \in F, \{j, l\} \notin F, \\ -a_{i}^{\dagger}a_{k}, & j = l, j \in F, \{i, k\} \notin F. \end{cases}$$
(5)

Moreover, this ability to neglect creation or annihilation operator with subscript from  $\{1, 2, 7, 8\}$ , along with the ability to neglect two-body operators containing an odd number of modes in F, allows us to relabel our orbital set 1 to 8, corresponding to spin-orbitals:  $\{3 \uparrow, 4 \uparrow, 5 \uparrow, 6 \uparrow, 3 \downarrow, 4 \downarrow, 5 \downarrow, 6 \downarrow\}$ 

Using the parity basis and taking advantage of particle and spin conversation, the required qubit number can be further reduced [25]. In the parity basis:

$$a_j^{\dagger} = X_{j+1}^{\leftarrow} \otimes \frac{1}{2} (X_j \otimes Z_{j-1} - iY_j), \tag{6}$$

$$a_j = X_{j+1}^{\leftarrow} \otimes \frac{1}{2} (X_j \otimes Z_{j-1} + iY_j), \tag{7}$$

where

$$X_i^{\leftarrow} \equiv X_{n-1} \otimes X_{n-1} \otimes \dots \otimes X_{i+1} \otimes X_i, \quad n = 8.$$
(8)

This fermionic Hamiltonian can now be mapped to an 8local Hamiltonian represented as a weighted sum of tensor products of Pauli matrix  $\{I_i, X_i, Y_i, Z_i\}$ , which almost preserves the ground state energy value. The new Hamiltonian in the electronic occupation number basis set can be mapped to the parity basis set as:

$$|f_1 f_2 \cdots f_8\rangle \rightarrow |q_1\rangle \otimes |q_2\rangle \otimes \cdots \otimes |q_8\rangle, \qquad (9)$$

where

$$q_i = \sum_{k=1}^{i} f_k \mod 2 \in \{0, 1\}.$$
 (10)

Here  $f_k$  represents the number of electrons occupying the *k*th spin-orbital, and  $q_k$  represents the sum of electron numbers from 1st to *k*th spin-orbital.

We can now assume that half of the left 6 electrons are spin-up and the other half are spin-down. If this is the case,  $|q_4\rangle = |1\rangle$ , and  $|q_8\rangle = |0\rangle$ , which means only  $Z_4, I_4, Z_8, I_8$  will apply on these states [26]. Since  $Z_4 |$  $q_4\rangle = - |q_4\rangle, Z_8 | q_8\rangle = |q_8\rangle$ , all  $Z_4, Z_8$  can be substituted by  $-I_4$  and  $I_8$ , with this assumption we can now reduce this problem to a 6-local Hamiltonian.

#### 3. Methods of simulation

After the parity transformation and simplifications made above we now have a reduced 6-local Hamiltonian describing H<sub>2</sub>O in the form:  $H = \sum_{i=1}^{L} \alpha_i h_i$ , where { $\alpha_i$ } is a set of coefficients, and { $h_i$ } is a set of tensor products of the Pauli matrices { $I_i, X_i, Y_i, Z_i$ }. Method A,B,C tries to evolve quantum system state by approximating the propagator  $e^{-iHt}$ , and then extract the ground state energy from the phase. Method D implements the Hamiltonian, H, directly into quantum circuit, and evaluate ground state energy by multiple measurements. Method E produces the ground state energies by iterations.

#### 3.1. Trotter phase estimate algorithm (Trotter-PEA)

For each term in a Hamiltonian, H, the propagator,  $e^{-i\alpha_i h_i t}$ , can be easily constructed in a circuit. However, since most of the time the set of  $h_i$  do not commute, the propagator cannot be implemented term by term: i.e.  $e^{-iHt} \neq \prod_{i=1}^{L} e^{-i\alpha_i h_i t}$ . The first-order Trotter–Suzuki decomposition [27–29] provides an easy way to decompose a propagator for the spin-type Hamiltonian given as a sum of non-commuting terms into a product of each non-commuting term exponentiated for a small time *t*:

$$U = \prod_{i=1}^{L} e^{-i\alpha_i h_i t} = e^{-iHt} + O(A^2 t^2).$$
(11)

Here  $A = \sum_{i=1}^{L} |\alpha_i|$ , and we have an error of order  $O(A^2t^2)$ . Here we don't consider time slicing as the original Trotter–Suzuki decompositoin does, as *t* can be adjusted to be as small as necessary for error control. This method requires only multi-qubit rotations, and therefore *U* can be implemented easily on a state register.

After *U* is obtained PEA can be applied to extract the phase. We can use extra ancilla qubits to achieve wanted accuracy by iterative measurements [10,30,31]. We call this PEA based on first-order Trotter–Suzuki decomposition Trotter-PEA.

Higher order Trotter–Suzuki decompositions are also available, however they have more complicated formulations, especially for order higher than 2. Here we only discuss first-order case for simplicity. For simulation, a forward iterative PEA [18] – which estimates the phase



**Figure 1.** Forward iterative PEA circuit with initial state  $|0\rangle | \psi\rangle_s$ . Here  $|\psi\rangle_s$  is the ground state of the Hamiltonian, *H* is the Hadamard gate, *U* is the approximate propagator and  $R_z(-(\pi/2))$  is a Z rotation gate.

starting from the most significant bit-can be used to save more time. The circuit for the forward iterative PEA is shown in Figure 1 which needs only 1 qubit for measurement. Then the generated state before the measurement is:

$$\frac{1 + e^{i2\pi(0.\phi_{k+1}\phi_{k+2}\cdots - 0.01)}}{2} e^{i(\pi/4)} |0\rangle |\psi\rangle_{s} + \frac{1 - e^{i2\pi(0.\phi_{k+1}\phi_{k+2}\cdots - 0.01)}}{2} e^{i(\pi/4)} |1\rangle |\psi\rangle_{s}.$$
 (12)

Note decimals above are in binary. It can be checked if the measurement qubit has a greater probability of output 1,  $\phi_{k+1} = 1$ , otherwise  $\phi_{k+1} = 0$ . Then the ground state energy can be calculated as  $E = -2\pi \times 0.\phi_1\phi_2\phi_3...$ 

# **3.2.** Direct implementation of hamiltonian in first-order (Direct-PEA (1st order))

It was proposed [17] that given a Hamiltonian H and large  $\kappa$  we can construct an approximated unitary operator U such that:

$$U = I - i\frac{H}{\kappa}, \quad \kappa \gg \sum_{i=1}^{L} |\alpha_i| \ge ||H||.$$
(13)

If  $|\psi\rangle_s$  is an eigenvector of *H* and *E* is the corresponding eigenvalue, then:

$$U \mid \psi \rangle_{s} = \left( I - i \frac{H}{\kappa} \right) \mid \psi \rangle_{s}$$
  
$$\approx e^{-i(H/\kappa)} \mid \psi \rangle_{s} = e^{-i(E/\kappa)} \mid \psi \rangle_{s}.$$
(14)

The eigenvalue of  $|\psi\rangle_s$  would be encoded directly in the approximate phase. This is the motivation behind directly implementing the Hamiltonian in quantum simulation.

To implement this non-unitary matrix U, we can enlarge the state space and construct a unitary operator  $U_r$  [13]. Rewrite U as:

$$U = I - \frac{i}{\kappa} \sum_{j=1}^{L} \alpha_j h_j = \sum_{j=0}^{L} \beta_j V_j,$$
 (15)

in which  $\beta_j \ge 0$  and  $V_j$  is unitary. By introducing a *m*-qubit ancilla register, where  $m = \lceil \log_2 L \rceil$ , we can construct a multi-control gate, *V*, such that:

$$V \mid j\rangle_a \mid \psi\rangle_s = \mid j\rangle_a V_j \mid \psi\rangle_s. \tag{16}$$

Define  $\beta_j = 0$  when  $L < j \le 2^m$  and *B* as a unitary operator that acts on ancilla qubits as:

$$B \mid 0 \rangle_a = \frac{1}{\sqrt{s}} \sum_{j=0}^{2^m} \sqrt{\beta_j} \mid j \rangle_a, \ s = \sum_{j=0}^{2^m} \beta_j.$$
(17)

Define  $U_r$  and  $\Pi$  such that:

$$U_r = (B^{\dagger} \otimes I^{\otimes n}) V(B \otimes I^{\otimes n}), \tag{18}$$

$$\Pi = |0\rangle_a \langle 0|_a \otimes I^{\otimes n}.$$
<sup>(19)</sup>

Apply  $U_r$  on input state  $|0\rangle_a |\psi\rangle_s$ :

$$\begin{aligned} U_r \mid 0 \rangle_a \mid \psi \rangle_s &= (B^{\dagger} \otimes I^{\otimes n}) V(B \otimes I^{\otimes n}) \mid 0 \rangle_a \mid \psi \rangle_s \\ &= (B^{\dagger} \otimes I^{\otimes n}) V \frac{1}{\sqrt{s}} \sum_{j=0}^{2^m} \sqrt{\beta_j} \mid j \rangle_a \mid \psi \rangle_s \\ &= (B^{\dagger} \otimes I^{\otimes n}) \frac{1}{\sqrt{s}} \sum_{j=0}^{2^m} \sqrt{\beta_j} \mid j \rangle_a V_j \mid \psi \rangle_s \\ &= \Pi (B^{\dagger} \otimes I^{\otimes n}) \frac{1}{\sqrt{s}} \sum_{j=0}^{2^m} \sqrt{\beta_j} \mid j \rangle_a V_j \mid \psi \rangle_s \\ &+ (I^{\otimes m+n} - \Pi) (B^{\dagger} \otimes I^{\otimes n}) \frac{1}{\sqrt{s}} \\ &\times \sum_{j=0}^{2^m} \sqrt{\beta_j} \mid j \rangle_a V_j \mid \psi \rangle_s \\ &= (B \mid 0 \rangle_a \langle 0 \mid_a)^{\dagger} \frac{1}{\sqrt{s}} \sum_{j=0}^{2^m} \sqrt{\beta_j} \mid j \rangle_a V_j \mid \psi \rangle_s \\ &+ \sum_{j=1}^{j=2^m} \mid j \rangle_a \mid u_j \rangle_s \\ &= \frac{1}{s} \mid 0 \rangle_a U \mid \psi \rangle_s + \mid \Phi_1^{\perp} \rangle, \end{aligned}$$

where  $|\Phi_1^{\perp}\rangle$  is orthogonal to  $|0\rangle_a |\psi\rangle_s$ . Then the approximated unitary operator *U* is implemented by unitary operator  $U_r$ , which can be seen in Figure 2.Since  $\kappa \gg ||H|| \ge E$ , energy of eigenstate  $|\psi\rangle_s$  is successfully



**Figure 2.** Gate  $U_r$  indirect PEA circuit, gates V and B are shown in Equations (16) and (17).

implemented in phase:

$$U_{r} |0\rangle_{a} |\psi\rangle_{s} = \frac{1 - i\frac{E}{\kappa}}{s} |0\rangle_{a} |\psi\rangle_{s} + |\Phi_{1}^{\perp}\rangle$$
  
$$= \frac{\sqrt{1 + \frac{E^{2}}{\kappa^{2}}}}{s} e^{-i\tan^{-1}(E/\kappa)} |0\rangle_{a} |\psi\rangle_{s} + |\Phi_{1}^{\perp}\rangle$$
  
$$= p e^{-i\tan^{-1}(E/\kappa)} |0\rangle_{a} |\psi\rangle_{s} + \sqrt{1 - p^{2}} |\Phi^{\perp}\rangle.$$
  
(21)

Here *p* is defined by  $\sqrt{1 + E^2/\kappa^2}/s$ , and  $| \Phi^{\perp} \rangle$  is normalised.

This  $U_r$  gate would then be used for PEA or iterative PEA process. For an accurate output, p is required to be as close to 1 as possible. Using oblivious amplitude amplification [32], we can amplify that probability without affecting phase. Define the operator  $U_0 = 2 \mid$  $0\rangle_a \langle 0_a - I^{\otimes m}$  and rotational operator:

$$Q = U_r(U_0 \otimes I^{\otimes n}) U_r^{\dagger}(U_0 \otimes I^{\otimes n}).$$
(22)

Iterating this operator N times, we can achieve  $U_q = Q^N U_r$  which brings p close to 1 by performing rotations within the space  $span\{|0\rangle_a | \psi\rangle_s, |\Phi^{\perp}\rangle\}$ . The details are in Supplementary Materials. Take the same circuit and the same procedure in Trotter-PEA, except replacing U by  $U_q$ , we are able to get ground state energy of water molecule.

# 3.3. Direct implementation of hamiltonian in second-order (Direct-PEA (2nd order))

Propagator  $e^{-iHt}$  can also be approximated up to second order [18]:

$$U = I - iHt - \frac{H^2 t^2}{2} = e^{-iHt} + O((At)^3).$$
 (23)

When At is very small, U would be a good approximation. Since U is non-unitary, we have to construct a unitary operator  $U_{r2}$  to implement it into a quantum circuit. With  $U_r$  in method B,  $B_2$  defined with the property:

$$B_2 \mid 00\rangle = \frac{\sqrt{t} \mid 00\rangle + \mid 01\rangle + \frac{t}{\sqrt{2}} \mid 10\rangle}{\sqrt{1 + t + \frac{t^2}{2}}}, \qquad (24)$$



**Figure 3.** Gate  $U_{r2}$  insecond-order direct PEA circuit, with  $B_2$  and *P* defined in Equations (24) and (25).

and gate *P* constructed as:

$$P = \begin{bmatrix} I^{\otimes n} & 0 & 0 & 0\\ 0 & 0 & I^{\otimes m} & 0\\ 0 & I^{\otimes m} & 0 & 0\\ 0 & 0 & 0 & I^{\otimes n} \end{bmatrix}.$$
 (25)

We can construct  $U_{r2}$  as in Figure 3:which satisfies:

$$\begin{aligned} U_{r2} \mid 00 \rangle \mid 0 \rangle_{a} \mid \psi \rangle_{s} \\ &= \frac{1 - i\frac{Et}{A} + \frac{E^{2}t^{2}}{2A^{2}}}{1 + t + \frac{t^{2}}{2}} \mid 00 \rangle \mid 0 \rangle_{a} \mid \psi \rangle_{s} + \sum_{j=1}^{2^{m+2}} \mid j \rangle \mid v_{j} \rangle_{s} \\ &= \frac{\sqrt{1 + \frac{E^{4}t^{4}}{4A^{2}}}}{1 + t + \frac{t^{2}}{2}} e^{-i\tan^{-1}((Et/A)/(1 + E^{2}t^{2}/2A))} \\ &\mid 00 \rangle \mid 0 \rangle_{a} \mid \psi \rangle_{s} + \mid \Psi_{1}^{\perp} \rangle. \end{aligned}$$
(26)

In the formula,  $A = \sum_{i=1}^{2^m-1} \beta_i = \sum_{i=1}^{L} |\alpha_i| \ge |E|$ , and  $|\Psi_1^{\perp}\rangle$  is perpendicular to  $|00\rangle |0\rangle_a |\psi\rangle_s$ . Just as in last section, we can rotate the final state to make the proportion of  $|00\rangle |0\rangle_a |\psi\rangle_s$  as close to 1 as possible. Then we can apply PEA or iterative PEA to get the phase,  $-\tan^{-1}((Et/A)/(1+E^2t^2/2A))$ , which leads to ground state energy corresponding to ground state  $|\psi\rangle_s$ .

#### 3.4. Direct measurement of Hamiltonian

Another way to calculate the ground state energy is by direct measurement after implementing a given Hamiltonian as a circuit. Since Direct-PEA (1st order) method has already introduced a way to implement non-unitary matrix U into circuit, Hamiltonian implementation is straightforward. We can just replace U in method B by  $U' = H = \sum_{j=1}^{L} \alpha_j h_j$ , and obtain  $U'_r$  such that:

$$U'_{r} |0\rangle_{a} |\psi\rangle_{s} = \frac{1}{s'} |0\rangle_{a} U'_{r} |\psi\rangle_{s} + |\Phi_{1}^{'\perp}\rangle$$
$$= \frac{E}{A} |0\rangle_{a} |\psi\rangle_{s} + |\Phi_{1}^{'\perp}\rangle.$$
(27)

By measuring ancilla qubits multiple times, we can get the energy of the ground state  $|\psi\rangle_s$  by multiplying *A* by the square root of probability of getting all 0s.



**Figure 4.** Circuit for state preparation and corresponding energy evaluation.  $G(\theta_i)$  is entangling gate, in this paper we are taking the gate like Figure 5.  $U(\varphi_k)$  is an arbitrary single-qubit rotation and is equal to  $R_z(\varphi_{k,1})R_x(\varphi_{k,2})R_z(\theta_{k,3})$  with parameters  $\varphi_{k,1},\varphi_{k,2}$  and  $\varphi_{k,3}$  that can be manipulated. By increasing the number of layers, *d*, of our circuit, we are able to produce more complex states.



**Figure 5.** Example entangling circuit  $G(\theta_i)$  for 4-qubit system. There are 12 arbitrary single-qubit gates  $U_j$ , a simplified written way for  $U(\theta_{i,j})$ , which is  $R_z(\theta_{i,j,1})R_x(\theta_{i,j,2})R_z(\theta_{i,j,3})$  with parameters  $\theta_{i,j,1}, \theta_{i,j,2}$  and  $\theta_{i,j,3}$  that can be manipulated. Each 2 qubits are entangled sequentially. Entangling gate  $G(\theta_i)$  for *n*-qubit system is similar to this gate, but then it has n(n-1) arbitrary single-qubit gates and  $\theta_i$  has 3n(n-1) parameters.

This method can also be used for non-hermitian Hamiltonians. If now the eigenvalue for  $|\psi\rangle_s$  is a complex number  $E = |E|e^{i\theta}$ , by replacing U by U' = H in method B, we would have:

$$U'_{r} |0\rangle_{a} |\psi\rangle_{s} = \frac{|E|e^{i\theta}}{A} |0\rangle_{a} |\psi\rangle_{s} + |\Phi_{1}^{'\perp}\rangle, \quad (28)$$

and can obtain |E| through measurements. Then by replacing U by U'' = (|E|/A)I + H in method B, we would have:

$$U_r'' \mid 0\rangle_a \mid \psi\rangle_s = \frac{\mid E \mid}{A} (1 + e^{i\theta}) \mid 0\rangle_a \mid \psi\rangle_s + \mid \Phi_1^{'\perp}\rangle,$$
(29)

and can measure the absolute value of  $(|E|/A)(1 + e^{i\theta})$ , which is  $2(|E|/A) \cos \theta$ . This helps determine the phase of a complex eigenenergy.

#### 3.5. Variational quantum eigensolver

Recently the variational quantum eigensolver method has been put forward by Aspuru-guzik and coworkers to calculate the ground state energies [19–22,33], which is a hybrid method of classical and quantum computation. According to this method, an adjustable quantum circuit is constructed at first to generate a state of the system. This state is then used to calculate the corresponding energy under the system's Hamiltonian. Then by a classical optimisation algorithm, like Nelder–Mead method, parameters in circuit can be adjusted and the generated state will be updated. Finally, the minimal energy will be obtained. The detailed circuit for the quantum part of our algorithm

is shown in Figure 4. To make the expression more clear, we represent parameters in vector form, as follows:  $\boldsymbol{\theta} = (\theta_1, \theta_2, \dots, \theta_D), \theta_i = (\theta_{i,0}, \theta_{i,1}, \dots, \theta_{i,11}), \theta_{i,j} =$  $(\theta_{i,j,1}, \theta_{i,j,2}, \theta_{i,j,3}), \varphi = (\varphi_1, \varphi_2 \dots, \varphi_n), \varphi_k = (\varphi_{k,1}, \varphi_{k,2}, \varphi_{k,2})$  $\varphi_{k,3}$ ). We are using d layers of gate  $G(\theta_i)$  in Figure 4 to entangle all qubits together. Here we introduce a hardware-efficient  $G(\boldsymbol{\theta}_i)$ , and we call this method Pairwise VQE. The example gate of  $G(\boldsymbol{\theta}_i)$  for 4 qubits is shown in Figure 5. The entangling gate for 6-qubit system H<sub>2</sub>O is similar: every 2 qubits are modified by singlequbit gates and entangled by CNOT gate. By selecting initial value of all  $\theta_i$  and  $\varphi_k$ , system state can be prepared by *d* layers  $G(\boldsymbol{\theta}_i)$  gates and arbitrary single gates  $U(\boldsymbol{\varphi}_i)$ . Then average value of each term in Hamiltonian H,  $\langle h_i \rangle$ , can be evaluated by measuring qubits many times after going through gates like *I* or  $R_{x_i}(\pi/2)$  or  $R_{y_i}(-(\pi/2))$ . For example, if  $h_i = I_0 X_1 Y_2 Z_3$ , then

$$\begin{split} \langle h_j \rangle &= \langle I_0 X_1 Y_2 Z_3 \rangle_{\psi} = \langle \psi I_0 X_1 Y_2 Z_3 \mid \psi \rangle \\ &= \left( \langle \psi R_{y_1} \left( \frac{\pi}{2} \right) R_{x_2} \left( -\frac{\pi}{2} \right) \right) I_0 \left( R_{y_1} \left( -\frac{\pi}{2} \right) X_1 \\ &R_{y_1} \left( \frac{\pi}{2} \right) \right) \left( R_{x_2} \left( \frac{\pi}{2} \right) Y_2 R_{x_2} \left( -\frac{\pi}{2} \right) \right) \\ &Z_3 \left( R_{y_1} \left( -\frac{\pi}{2} \right) R_{x_2} \left( \frac{\pi}{2} \right) \mid \psi \rangle \right) \\ &= \langle I_0 Z_1 Z_2 Z_3 \rangle_{\psi'}, \quad \text{where } \mid \psi' \rangle \\ &= R_{y_1} \left( -\frac{\pi}{2} \right) R_{x_2} \left( \frac{\pi}{2} \right) \mid \psi \rangle, \end{split}$$

So we can let the quantum state after  $U(\varphi_j)$  go through gates  $R_{y_1}(-(\pi/2))$  and  $R_{x_2}(\pi/2)$  and then measure the



**Figure 6.** Ground State Energy Curve for H<sub>2</sub>O, as a function of the bond length O–H in a.u. for (a) the Trotter-PEA, (b) the Direct-PEA (1st order), (c) the Direct-PEA ( $2^{nd}$  order) and (d) Direct Measurement method ( $1.6 \times 10^8$  measurements), compared with the exact diagonalisation. Errors are shown in the window of each figure. One thing to mention is that we can not tell whether one method have better property over another directly from these figures, because they have different parameters, gates etc. For comparison, we have to turn to gate complexity analysis in Table 1.

result state multiple times to get  $\langle h_j \rangle$ . The energy corresponding to the state can be obtained by  $\langle H \rangle (\theta, \varphi) = \sum_{j=1}^{L} \alpha_j \langle h_j \rangle (\theta, \varphi)$ . Then  $\theta$  and  $\varphi$  can be updated by classical optimisation method and  $\langle H \rangle (\theta, \varphi)$  can reach the minimal step by step.

## 4. Results and method comparison

The Hamiltonian of the water molecule is calculated for O-H bond lengths ranging from 0.5 a.u. to 2.9 a.u., using the methods introduced in Section2. This Hamiltonian is used in all five of the methods discussed within this paper. For the methods A-D, the input state of system is the ground state of the H<sub>2</sub>O molecule. For each of these methods, the resulting ground state energy curve can be calculated to arbitrary accuracy (for details of error analysis see Appendix 2). The results from each method is compared with result from a direct diagonalisation of the Hamiltonian, as shown below. From Figure 6 it can be seen that all of these methods are effective in obtaining the ground state energy problem of the water molecule.

We also use method E (Pairwise VQE) to obtain the ground state energy. These results can be seen in Figure 7. Energy convergence at 1.9 a.u. can be seen in Figure 7(a) and the ground state energy curve calculated by this method is in Figure 7(b). In this simulation, *d* is selected to be 1, and  $G(\theta_i)$  is constructed as described above, and it can already give a very accurate result. This shows Pairwise VQE a very promising method for solving electronic structure problems. Furthermore, Pairwise VQE has only  $O(n^2d)$  gate complexity and doesn't require initial input of the ground state, which makes it more practical for near-term applications on a quantum computer.

Qubit requirement, gate complexity and number of measurements of different methods are analysed in Appendix 3 and shown in Table 1. When counting gate complexities, we decompose all gates into single qubit gates and CNOT gates. While Pairwise VQE needs only n qubits, the other methods require extra number of qubits. In terms of gate scaling, Pairwise VQE also needs the least gates, which enables it to better suit the applications on near and intermediate term quantum computers.



**Figure 7.** Result from Pairwise VQE using the entangling gates in Figure 5. We take  $|0\rangle_s$  as initial input, d = 1 layer and use Nelder–Mead algorithm for optimisation. (a) Convergence of ground state energy of H<sub>2</sub>O for fixed O–H bond length = 1.9 a.u., as number of iterations increases. The lines for exact ground state energy and for the limit almost overlap and (b) Ground state energy curve for H<sub>2</sub>O, as a function of O–H bond length in a.u. for variational guantum eigensolver. Errors are shown in the window of the figure.

Table '	<ol> <li>Complexit</li> </ol>	y of different me	thods
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Method	Qubits requirement	Gate complexity	Number of measurements
Trotter-PEA	<i>O</i> ( <i>n</i> )	$O(\frac{n^5}{(\epsilon/A)^2})$	0(1)
Direct-PEA(1st order)	<i>O</i> ( <i>n</i> )	$O(\frac{n^5}{(\epsilon/A)^{2.5}})$	<i>O</i> (1)
Direct-PEA(2nd order)	<i>O</i> ( <i>n</i> )	$O(\frac{n^5}{(\epsilon/A)^{1.3}})$	<i>O</i> (1)
Direct measurement	<i>O</i> ( <i>n</i> )	$O(n^5)$	$O(\frac{E^2}{\epsilon^2})$
Pairwise VQE	n	$O(n^2d)$	$O(\frac{A^2n^{\delta}}{\epsilon^2}N_{\text{iter}})$

Notes: *n* is the number of qubits for molecular system, 6 for water in this paper.  $A = \sum_{i=1}^{l} |\alpha_i|$  can serve as the scale of energy. E is the exact value of ground energy.  $\epsilon$  is the accuracy of energy we want to reach. *d* is the number of layers we used in Pairwise VQE. *N*<sub>iter</sub> is the number of iterations for optimisation in Pairwise VQE. See Appendix 3 for details.

Among the remaining four methods, Direct Measurement requires less number of gates than the others. PEAtype methods have an advantage that they can give an accurate result under only O(1) measurements. However, they need more qubits compared with the previous two methods and demands many more gates if smaller error is required. Due to huge gate complexity, these PEAtype algorithms would be put into practice only when the decoherence problem has been better solved. Among these three PEA-based methods, in terms of the gate complexity, Direct-PEA(2nd order) requires less number of gates than the traditional Trotter-PEA and Direct-PEA(1st order) which is proved in Appendix 3. One more thing to mention is that here the second quantisation form Hamiltonian is based on STO-3G, so there are  $O(n^4)$  terms. If a more recent dual form of plane wave basis [15] is used, the number of terms can be reduced to  $O(n^2)$ , and the asymptotic scaling in Table 1 would also be reduced. To be specific, for PEA-type methods, upper bounds of gate complexities would be proportional to  $n^3$  rather than  $n^5$ , and Number of Measurements for Pairwise VQE would be proportional to  $n^4$  rather than  $n^8$ . As can be seen, these reductions wouldn't influence the comparison made above.

# 5. Excited states and resonances

All the aforementioned methods can also be applied for the excited state energy calculation. For PEA-type methods and Direct Measurement method, it can be simply done by replacing the input system state by an excited state. The complexity for the calculation is the same. The energy accuracies for excited states are also similar to that for the ground state. For VQE, a recent publication [34,35] presents a quantum subspace expansion algorithm (QSE) to calculate excited state energies. They approximate a 'subspace' of low-energy excited states from linear combinations of states of the form  $O_i | \psi \rangle_s$ , where  $|\psi\rangle_s$  is the ground state determined by VQE and  $O_i$  are chosen physically motivated quantum operators. By diagonalising the matrix with elements  $\langle \psi_s O_i^{\dagger} H O_j |$  $\psi_s$  calculated by VQE, one is able to find the energies of excited states.

Figure 8 shows the simulation of the first six excited states' energy curves of the water molecule from our 6qubit Hamiltonian, calculated by PEA-type methods and Direct Measurement method. It can be seen that the 5th excited energy curve indicates a shape resonance phenomenon, which can be described by a non-Hermitian Hamiltonian with complex eigenvalues. The life time of the resonance state is associated with the imaginary part of the eigenvalues. In this way, to solve the resonance problem, we can seek to solve the eigenvalues of non-Hermitian Hamiltonians.



**Figure 8.** Excited states' energy curves for H<sub>2</sub>O, as a function of the bond length O–H in a.u.. Markers with different colours represent data points calculated from different methods. Only a few points for each method are drawn for illustration. Energy curves in different line styles are calculated from exact diagonalisation of Hamiltonian matrix.

Some work has been done on this track to solve the resonance problem by quantum computers. By designing a general quantum circuit for non-unitary matrices, Daskin et al. [36] explored the resonance states of a model non-Hermitian Hamiltonian. To be specific, he introduced a systematic way to estimate the complex eigenvalues of a general matrix using the standard iterative phase estimation algorithm with a programmable circuit design. The bit values of the phase qubit determines the phase of eigenvalue, and the statistics of outcomes of the measurements on the phase qubit determines the absolute value of the eigenvalue. Other approaches for solving complex eigenvalues can also be applied for this resonance problem. For example, Wang et al. [37] proposed a measurement-based quantum algorithms for finding eigenvalues of non-unitary matrices. Terashima and Ueda [38] introduced a universal non-unitary quantum circuit by using a specific type of one-qubit nonunitary gates, the controlled-NOT gate, and all one-qubit unitary gates, which is also useful for finding the eigenvalues of a non-hermitian Hamiltonian matrix.

Method D in Section 3 can also be used for solving complex eigenvalues and the complexity is polynomial in system size. After applying complex-scaling method [39] to water molecule's Hamiltonian and obtaining a non-Hermitian Hamiltonian, we can make enough quantum measurements to get an accurate resonance width  $\Gamma$ , which is actually the imaginary part of Hamiltonian's eigenvalue [40]. Another easier way to solve this resonance problem is, we can first choose proper *a* and *J* to fit the potential energy in a widely studied Hamiltonian [41–43]:

$$H(x) = \frac{p^2}{2} + \left(\frac{x^2}{2} - J\right)e^{-ax^2}$$
(30)

to our energy curve. Then by complex-scaling method, the internal coordinates of the Hamiltonian is dilated by a complex factor  $\eta = \alpha e^{-i\theta}$  such that  $H(x) \rightarrow H(x/\eta) \equiv H_{\eta}(x)$ . We can solve the complex eigenvalue of  $H_{\eta}(x)$  by the method D or using our previous method [36].

#### 6. Conclusion

In this study we have compared several recently proposed quantum algorithms when used to compute the electronic state energies of the water molecule. These methods include first-order Trotter-PEA method based on the first-order Trotter decomposition, first- and secondorder Direct-PEA methods based on direct implementation of the truncated propagator, Direct Measurement method based on direct implementation of the Hamiltonian and Pairwise PEA method, a VQE algorithm with a designed ansatz.

After deriving the Hamiltonian of the water molecular using the STO-3G basis set, we have explained in detail how each method works and derived their qubit requirements, gate complexities and measurement scalings. We have also calculated the ground state energy of the water molecular and shown the ground energy curves from all five methods. All methods are able to provide an accurate result. We have compared these methods and concluded that the second-order Direct-PEA provides the most efficient circuit implementations in terms of gate complexity. With large scale quantum computation, the second-order direct method seems to better suit large molecule systems. In addition, since Pairwise VQE requires the least qubit number, it is the most practical method for near-term applications on the current available quantum computers.

Moreover, we have applied our PEA-type methods and Direct Measurement method to solve excited state energy curves for water molecule. The fifth excited state energy curve implies shape resonance. We have introduced recent work on quantum algorithms for solving the molecular resonance problems and given two possible ways to solve the water molecule resonance properties, including our Direct Measurement method which is able to solve the problem efficiently.

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No potential conflict of interest was reported by the authors.

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### **Appendices**

# Appendix 1. H<sub>2</sub>O Hamiltonian at equilibrium

**Table A1.** Spin-type Hamiltonian of the water molecule at equilibrium when O–H is 1.9 a.u.

	-72.008089	IIIIZ	0.373979	IIIIXX	-0.050755
IIIIYY	0.113535	IIIIZI	0.002526	IIIIZZ	0.779273
IIIZII	-0.771553	IIIZIZ	0.043092	IIIZXX	0.113535
IIIZYY	-0.050755	IIIZZI	0.785287	IIIZZZ	-0.030367
IIXIIX	0.009295	IIXIXI	0.000158	IIXIZX	-0.009295
IIXZXZ	-0.000158	IIZIII	-0.373979	IIZIIZ	-0.148141
IIZIYY	-0.011744	IIZIZZ	-0.146285	IIZZII	0.141059
IIZZXX	-0.011744	IIZZZI	-0.136887	IXIIIX	0.000158
IXIIXI	0.013400	IXIIZX	-0.000158	IXIZXZ	-0.013400
IXXIII	-0.050755	IYYIII	0.113535	IYYIIZ	0.011744
IYYIYY	0.019371	IYYIZZ	0.031747	IYYZII	-0.011216
IYYZXX	0.019371	IYYZZI	0.031561	IZIIII	-0.002526
IZXIIX	0.009295	IZXIXI	0.000158	IZXIZX	-0.009295
IZXZXZ	-0.000158	IZZIII	0.779273	IZZIIZ	0.146285
IZZIYY	0.031747	IZZIZZ	0.220040	IZZZII	-0.154863
IZZZXX	0.031747	IZZZZI	0.179396	XIIXII	0.012412
XIIXXX	-0.007950	XIIXZI	0.012412	XIIYXY	0.007950
XXXXII	-0.007950	XXXXXX	0.018156	XXXXZI	-0.007950
XXXYXY	-0.018156	XXZXXZ	-0.006979	XXZYYI	0.006979
XZIXII	-0.012412	XZIXXX	0.007950	XZIXZI	-0.012412
XZIYXY	-0.007950	YXYXII	0.007950	YXYXXX	-0.018156
YXYXZI	0.007950	YXYYXY	0.018156	YYIXXZ	-0.006979
YYIYYI	0.006979	ZIIIII	0.771553	ZIIIIZ	0.141059
ZIIIYY	0.011216	ZIIIZZ	0.154863	ZIIZII	-0.154860
ZIIZXX	0.011216	ZIIZZI	0.146877	ZIZIII	0.043092
ZXXIII	-0.113535	ZXXIIZ	-0.011744	ZXXIYY	-0.019371
ZXXIZZ	-0.031747	ZXXZII	0.011216	ZXXZXX	-0.019371
ZXXZZI	-0.031561	ZXZIIX	-0.000158	ZXZIXI	-0.013400
ZXZIZX	0.000158	ZXZZXZ	0.013400	ZYYIII	0.050755
ZZIIII	0.785287	ZZIIIZ	0.136887	ZZIIYY	0.031561
ZZIIZZ	0.179396	ZZIZII	-0.146877	ZZIZXX	0.031561
ZZIZZI	0.189343	ZZZIII	0.030367		

Notes: There are 95 terms, and listed are each operator and corresponding coefficient.  $X_i Y_i Z_i I$  stand for the spin matrices  $\sigma^x, \sigma^y, \sigma^z$  and the identity operator on a single qubit subspace.

### **Appendix 2. Error analysis**

#### A.2.1 Trotter PEA

The Trotter decomposition is

$$e^{-iHt} = \prod_{i=1}^{L} e^{-i\alpha_i h_i t} + O(A^2 t^2).$$
 (A1)

Suppose our input is an eigenstate of  $|\varphi\rangle_s$  of the Hamiltonian and has  $H |\varphi\rangle_s = E |\varphi\rangle_s$ , then:

$$\begin{split} \prod_{i=1}^{L} e^{-i\alpha_{i}h_{i}t} \mid \varphi \rangle_{s} \\ &= (e^{-iHt} - O(A^{2}t^{2})) \mid \varphi \rangle_{s} \\ &= (e^{-iEt} - O(A^{2}t^{2})) \mid \varphi \rangle_{s} + O(A^{2}t^{2}) \mid \varphi^{\perp} \rangle \\ &= e^{-iEt}(1 - O(A^{2}t^{2})e^{iE}) \mid \varphi \rangle_{s} + O(A^{2}t^{2}) \mid \varphi^{\perp} \rangle \\ &= e^{-iEt}(1 - O(A^{2}t^{2}) - iO(A^{2}t^{2})) \mid \varphi \rangle_{s} + O(A^{2}t^{2}) \mid \varphi^{\perp} \rangle \\ &= (1 - O(A^{2}t^{2}))e^{-iEt}e^{i\tan^{-1}(O(A^{2}t^{2})/(1 - O(A^{2}t^{2})))} \mid \varphi \rangle_{s} \\ &+ O(A^{2}t^{2}) \mid \varphi^{\perp} \rangle \\ &= (1 - O(A^{2}t^{2}))e^{-i(Et + O(A^{2}t^{2}))} \mid \varphi \rangle_{s} + O(A^{2}t^{2}) \mid \varphi^{\perp} \rangle. \end{split}$$
(A2)

It should be noticed that in this equation,  $O(A^2 t^2)$  is an operator before being applied to  $|\varphi\rangle_s$ .

In this way, the possibility that we can measure the correct ground state energy is  $1 - O(A^2t^2)$ . After  $2^D$  gates, in which D represents the number of digits we want to measure by PEA, the probability of state  $|0\rangle |\psi\rangle_s$  should be still large. By setting the final coefficient to be  $1 - \frac{1}{8}$ , then:

$$(1 - O(A^2 t^2))^{2^D} = 1 - \frac{1}{8},$$
 (A3)

$$2^{-D} = O(A^2 t^2). (A4)$$

The error of the energy resulting from the phase is:  $\epsilon_1 = O(A^2t^2)$ . If we use PEA until *D* digits, the error of the energy resulting from PEA is:  $\epsilon_2 = O(2^{-D}/t) = O(A^2t)$ . Then totally we have an error:  $\epsilon = O(\epsilon_1 + \epsilon_2) = O(A^2t)$ .

Since the error for the first-order Trotter–Suzuki decomposition is:

$$e^{-iHt} - \prod_{i=0}^{L} e^{-(i\alpha_i h_i t/2)} \prod_{i'=L}^{0} e^{-(i\alpha'_i h_{i'} t/2)} = O(A^3 t^3), \quad (A5)$$

by a similar analysis the total error after PEA based on Trotter–Suzuki decomposition would be  $\epsilon = O(A^3 t^2)$ .

#### A.2.2 Direct PEA (first order)

From the main part, after gate  $U_r$ , we obtain:

$$U_r \mid 0\rangle_a \mid \psi\rangle_s = \frac{\sqrt{1 + \frac{E^2}{\kappa^2}}}{s} e^{-i\tan^{-1}(E/\kappa)} \mid 0\rangle_a \mid \psi\rangle_s + \mid \Phi_1^{\perp}\rangle$$
$$= \frac{\sqrt{1 + \frac{E^2}{\kappa^2}}}{1 + \frac{A}{\kappa}} e^{-i\tan^{-1}(E/\kappa)} \mid 0\rangle_a \mid \psi\rangle_s + \mid \Phi_1^{\perp}\rangle$$

$$= p e^{-i \tan^{-1}(E/\kappa)} |0\rangle_a |\psi\rangle_s + \sqrt{1-p^2} |\Phi^{\perp}\rangle$$
  
$$= \cos \theta e^{-i \tan^{-1}(E/\kappa)} |0\rangle_a |\psi\rangle_s + \sin \theta |\Phi^{\perp}\rangle.$$
  
(A6)

Here  $A = \sum_{i=1}^{2^m - 1} \beta_i = \sum_{i=1}^{L} |\alpha_i| \ge |E|, \qquad \theta = \arccos\left(\sqrt{1 + E^2/\kappa^2}/(1 + A/\kappa)\right).$ 

To increase the probability of  $|0\rangle_a |\psi\rangle_s$ , we use  $Q = U_r(U_0 \otimes I^{\otimes n})U_r^{\dagger}(U_0 \otimes I^{\otimes n})$  to do oblivious amplitude amplification:

$$Q^{N}U_{r} |0\rangle_{a} |\psi\rangle_{s} = (-1)^{N} \cos((2N+1)\theta) e^{-i\tan^{-1}(E/\kappa)} |0\rangle_{a} |\psi\rangle_{s}$$
$$+ \sin((2N+1)\theta) |\Phi^{\perp}\rangle$$
$$= p_{f} |0\rangle_{a} |\psi\rangle_{s} + \sqrt{1-p_{f}^{2}} |\Phi^{\perp}\rangle.$$
(A7)

The idea is, if  $\kappa$  is large,  $\sqrt{1 + E^2/\kappa^2}/(1 + A/\kappa) \approx 1/(1 + A/\kappa)$ , and  $\theta' = \cos^{-1}(1/(1 + A/\kappa)) \approx \theta$ . By choosing large N and  $\kappa$  to satisfy  $(2N + 1)\theta' = \pi$ , which means  $A/\kappa = 1/\cos(\pi/2N + 1) - 1$ , we are able to get  $\cos((2N + 1)\theta) \approx -1$ .

Since:

$$\theta - \theta' = \cos^{-1}\left(\frac{\sqrt{1 + \frac{E^2}{\kappa^2}}}{1 + \frac{A}{\kappa}}\right) - \cos^{-1}\left(\frac{1}{1 + \frac{A}{\kappa}}\right)$$
$$= \frac{\sqrt{2}}{4}\eta^2 \left(\frac{A}{\kappa}\right)^{3/2} + O\left(\left(\frac{A}{\kappa}\right)^{5/2}\right).$$
(A8)

In which  $\eta = |E/A| \le 1$ . Then after *N* rotations

$$|p_{f}| = |\cos((2N+1)\theta)|$$
  
=  $\cos((2N+1)(\theta'-\theta))$   
=  $1 - \frac{(2N+1)^{2}}{16}\eta^{4}\left(\frac{A}{\kappa}\right)^{3} + O\left(\left(\frac{A}{\kappa}\right)^{4}\right)$   
=  $1 - \frac{\pi^{6}}{2^{11}}\eta^{4}\frac{1}{N^{4}} + O\left(\frac{1}{N^{5}}\right).$  (A9)

This means if we set large enough *N*, and then set  $\kappa = A \cos(\pi/(2N+1))/(1 - \cos(\pi/(2N+1)))$ , we are able to amplify the probability of  $|0\rangle_a |\psi\rangle_s$  to be as close to 1 as we want.

Now we are taking  $U_q = Q^N U_r$  to encode the energy into the phase. For the next PEA step, if we would like *D* digit accuracy, we have to make sure after  $2^D$  gates of  $U_q$ , the probability of state  $|0\rangle_a |\psi\rangle_s$  is still large. To make the analysis easier, we set the final coefficient for that state  $1 - 1/2^3$ . Then the following formula should be satisfied:

$$|p_f|^{2^D} = 1 - \frac{1}{2^3} \tag{A10}$$

$$2^{-D} = \frac{\pi^6 \eta^4}{2^{11} \ln(\frac{8}{7})} \frac{1}{N^4} + O\left(\frac{1}{N^5}\right)$$
(A11)

$$D = \min\left\{\log_2\left(\frac{2^{11}\ln(\frac{8}{7})}{\pi^6\eta^4}\right) + 4\log_2 N\right\} \approx -1.81 + 4\log_2 N.$$
(A12)

Since D-digit output from PEA gives us the phase  $\varphi$  to approximate  $(1/2\pi) \tan^{-1}(-E/\kappa)$ , and the error of the phase is  $2^{-D}$ , we get the error of the energy to be:

$$\epsilon = \tan(2\pi * 2^{-D}) \times \kappa = \frac{\pi^5 \eta^4}{2^7 \ln(\frac{8}{7})} \frac{1}{N^2} + O\left(\frac{1}{N^3}\right)$$
$$\approx \frac{17.90 \eta^4 A}{N^2} \le \frac{17.90}{N^2} A.$$
(A13)

We can see that, by taking large N and set corresponding  $\kappa$  (which is also large), we are able to control the accuracy of PEA process.

#### A.2.3 Direct PEA (second order)

From the main part, after gate  $U_{2r}$ , we obtain:

$$U_{r2} | 00\rangle | 0\rangle_{a} | \psi\rangle_{s}$$

$$= \frac{\sqrt{1 + \frac{E^{4}t^{4}}{4A^{2}}}}{1 + t + \frac{t^{2}}{2}} e^{-i\tan^{-1}((Et/A)/(1 + E^{2}t^{2}/2A))}$$

$$| 00\rangle | 0\rangle_{a} | \psi\rangle_{s} + | \Psi_{1}^{\perp}\rangle$$

$$= pe^{-i\tan^{-1}((Et/A)/(1 + E^{2}t^{2}/2A))}$$

$$| 00\rangle | 0\rangle_{a} | \psi\rangle_{s} + \sqrt{1 - p^{2}} | \Psi^{\perp}\rangle$$

$$= \cos\theta e^{-i\tan^{-1}((Et/A)/(1 + E^{2}t^{2}/2A))}$$

$$| 00\rangle | 0\rangle_{a} | \psi\rangle_{s} + \sin\theta | \Psi^{\perp}\rangle. \quad (A14)$$

Here  $A = \sum_{i=1}^{2^m - 1} \beta_i = \sum_{i=1}^{L} |\alpha_i| \ge |E|, \theta = \cos^{-1} (\sqrt{1 + E^4 t^4 / 4A^2} / (1 + t + t^2 / 2)).$ 

Apply  $Q_2 = U_{2r}(U_0^+ \otimes I^{\otimes n})U_{2r}^{\dagger}(U_0^+ \otimes I^{\otimes n})$ , in which  $U_0^+ = 2 \mid 00 \rangle \mid 0 \rangle_a \langle 0_a \langle 00 - I^{\otimes m+2} \rangle$ , to do obivious amplitude amplification:

$$Q_{2}^{N}U_{2r} \mid 00 \mid 0_{a} \mid \psi_{s}$$

$$= (-1)^{N} \cos((2N+1)\theta) e^{-i\tan^{-1}((Et/A)/(1+E^{2}t^{2}/2A))}$$

$$\mid 0_{a} \mid \psi_{s} + \sin((2N+1)\theta) \mid \Psi^{\perp}{}_{a+s+2}$$

$$= p_{f} \mid 00 \mid 0_{a} \mid \psi_{s} + \sqrt{1-p_{f}^{2}} \mid \Psi^{\perp}{}_{a+s+2}.$$
(A15)

Let  $\theta' = \cos^{-1}(1/(1 + t + t^2/2))$  and choose large N and small t to satisfy  $(2N + 1)\theta' = \pi$ , which leads to

$$t = -1 + \sqrt{\frac{2}{\cos\frac{\pi}{2N+1}} - 1} = \frac{\pi^2}{8N^2} + O\left(\frac{1}{N^3}\right).$$
 (A16)

Then:

$$\theta - \theta' = \cos^{-1} \frac{\sqrt{1 + \frac{E^4 t^4}{4A^2}}}{1 + t + \frac{t^2}{2}} - \cos^{-1} \frac{1}{1 + t + \frac{t^2}{2}}$$
$$= \frac{\sqrt{2}}{16} \eta^4 t^{7/2} + O\left(t^{9/2}\right). \tag{A17}$$

In which  $\eta = |E/A| \le 1$ . Then after *N* rotations

$$\begin{aligned} |p_f| &= |\cos((2N+1)\theta)| \\ &= \cos((2N+1)(\theta'-\theta)) \\ &= 1 - \frac{(2N+1)^2}{16^2} \eta^8 t^7 + O(t^8) \\ &= 1 - \frac{\pi^{14}}{2^{27}} \eta^8 \frac{1}{N^{12}} + O\left(\frac{1}{N^{13}}\right). \end{aligned}$$
(A18)

This means if we set large enough N, and then set  $t = -1 + \sqrt{2/\cos(\pi/(2N+1)) - 1}$ , we are able to amplify the probability of  $|00\rangle |0\rangle_a |\psi\rangle_s$  to be as close to 1 as we want.

Now we are taking  $U_{q2} = Q_2^N U_{r2}$  to encode the energy into the phase, if we would like *D* digit accuracy, we have to make sure after  $2^D$  gates of  $U_{q2}$ , the probability of state  $|0\rangle_a | \psi\rangle_s$ is still large. By setting the final coefficient is  $1 - \frac{1}{2^3}$ , then the following formula should be satisfied:

$$|p_f|^{2^D} = 1 - \frac{1}{2^3} \tag{A19}$$

$$2^{-D} = \frac{\pi^{14} \eta^8}{2^{27} \ln(\frac{8}{7})} \frac{1}{N^{12}} + O\left(\frac{1}{N^{13}}\right)$$
(A20)

$$D = \min\left\{\log_2\left(\frac{2^{27}\ln(\frac{8}{7})}{\pi^{14}\eta^8}\right) + 12\log_2 N\right\}$$
  
\$\approx 0.974 + 12\log\_2 N.\$ (A21)

Since D-digit output from PEA gives us the phase  $\varphi$  to approximate  $-(1/2\pi) \tan^{-1}((Et/A)/(1+E^2t^2/2A))$  and the error of phase is  $2^{-D}$ , we get the error of the energy *E* to be:

$$\epsilon = \frac{\pi^{13} \eta^8}{2^{23} \ln \frac{8}{7}} \frac{A}{N^{10}} + O\left(\frac{A}{N^{11}}\right) \approx \frac{2.59 \eta^8 A}{N^{10}} \le \frac{2.59}{N^{10}} A.$$
(A22)

We can see that by taking large *N* and set corresponding small *t*, we are able to control the accuracy of PEA process.

#### A.2.4 Direct measurement

After applying the gate  $U'_r$ :

$$U'_{r} |0\rangle_{a} |\psi\rangle_{s} = \frac{E}{A} |0\rangle_{0} |\psi\rangle_{s} + |\Phi_{1}^{'\perp}\rangle.$$
 (A23)

We obtain eigenenergy of state  $|\psi\rangle_s$  by calculating probability of the wanted state:  $|0\rangle_a |\psi\rangle_s$ . The standard error of *E* by *X* measurements is:

$$\sigma = \frac{|E|}{\sqrt{X}} \sqrt{1 - \frac{E^2}{A^2}}.$$
 (A24)

#### **Appendix 3. Complexity**

#### A.3.1 Trotter PEA

We need n qubits for the state and at least 1 qubit for PEA process, so totally we need O(n) qubits.

If we measure the ground state energy to *D* bit precision, we need  $O(2^{D}Ln)$  standard gates to implement PEA, in which by saying standard gates we mean single qubit gates and CNOT gates. Since  $2^{D} = O(1/A^{2}t^{2}) = O(A^{2}/\epsilon^{2})$ , and for molecular

system,  $L = n^4$ , so the gate complexity of Trotter PEA would be  $O(n^5/(\epsilon/A)^2)$ .

For PEA based on the second-order Trotter–Suzuki decomposition, we still need  $O(2^{D}Ln)$  standard gates. Now  $2^{D} = O(1/A^{3}t^{3}) = O(A^{1.5}/\epsilon^{1.5})$ , so the gate complexity would be  $O(n^{5}/(\epsilon/A)^{1.5})$ .

#### A.3.2 Direct PEA (first order)

To do Direct PEA, we need *n* qubits to represent the system state and  $m = \lceil \log_2(L) \rceil$  qubits to represent the ancilla state. We also need at least 1 qubit for multi-control Toffoli gates in B gate construction [44]. Towards molecular system of  $L = O(n^4)$ , so the number of required qubits is O(n).

To meet properties of B, we can use Householder transformation and set it as:

$$B = I - \frac{2}{\langle u | u_a \rangle} | u \rangle \langle u_a, \qquad (A25)$$

where  $|u\rangle_a = B |0\rangle_a - |0\rangle_a$ . The complexity of constructing this gate has been analysed before [17,45–47]. Since Givens rotation  $G_{L-2,L-1}(\theta_{L-1})$  can nullify  $B_{0,L-1}$ , it can also nullify all  $B_{j,L-1}$  for  $j \neq L - 1$  and update  $B_{L-1,L-1}$  to 1 due to *B*'s special form. And  $G_{L-2,L-1}^T(\theta_{L-1})$  would nullify all  $B_{L-1,j}$  except  $B_{L-1,L-1}$ . For index smaller than L-1 but larger than 1, we can do the same thing. Finally we can choose  $G_{1,1}(\theta_1)$  to update last 4 elements of B and obtain an identity matrix. Thus we have:

$$G_{1,1}(\theta_1) \prod_{i=2}^{L-1} G_{i-1,i}(\theta_i) B \prod_{i=L-1}^2 G_{i-1,i}^T(\theta_i) = I, \qquad (A26)$$

$$B = \prod_{i=L-1}^{2} G_{i-1,i}^{T}(\theta_{i}) G_{1,1}^{T}(\theta_{1}) \prod_{i=2}^{L-1} G_{i-1,i}(\theta_{i}).$$
(A27)

In this way, *B* gate can be obtained as a product of 2L-3 Givens rotation matrices. Since each Givens rotation matrix can be achieved by at most *m m*-control Toffoli gates, which would cost  $O(m^2)$  standard gates [44,48] each, totally  $O(Lm^3) = O(L\log^3 n)$  gates are required. For *select*(*V*) gate, we need O((n + m)L) standard gates. In this way,  $U_r$  requires  $O(L\log^3 n + (n + m)L) = O(n^5)$  gates.  $U_0$  only needs O(m) standard gates, so Q also requires  $O(n^5)$  standard gates, which leads the gate complexity of  $U_q$  to be  $O(Nn^5)$ . Since  $N = O(1/(\epsilon/A)^{1/2})$ , PEA for *D* digit accuracy would result in a total of  $O(2^DNn^5) = O(n^5/(\epsilon/A)^{2.5})$  standard gates.

#### A.3.3 Direct PEA (second order)

We need *n* qubits to represent the system state,  $m' = \lceil \log_2(L) \rceil + 2$  qubits to represent the ancilla state. So the number of required qubits is still O(n) as the first-order direct PEA.

When constructing  $U_{r2}$ , gate  $U_r$  takes  $O(n^5)$  standard gates, gate P takes  $O(L) = O(n^4)$  standard gates,  $B_2$ ,  $B_2^{\dagger}$  and phase gate  $e^{-i(\pi/2)}$  only takes a small constant of standard gates. So the gate complexity of  $U_{r2}$  is still  $O(n^5)$ .  $Q_2$  also requires  $O(n^5)$ standard gates since  $U_0^{\pm}$  needs O(m) standard gates. Since  $N = O(1/(\epsilon/A)^{0.1})$ , PEA for D digit accuracy would result in a toltal of  $O(2^D Nn^5) = O(n^5/(\epsilon/A)^{1.3})$  standard gates.

# A.3.4 Direct measurement

The number of required qubits for Direct Measurement Method is the sum of system and ancilla qubits: O(n). Since only one  $U_r$  gate is enough, the compexity of the standard gates is  $O(n^5)$ . Since now the result of measurements is a binomial distribution, to measure the Energy *E* to accuracy(standard deviation)  $\epsilon$ , we have to make  $X = E^2/\epsilon^2$  measurements.

# A.3.5 Variational quantum eigensolver

The number of qubits required for Pairwise VQE is n, and the gate complexity is  $O(n^2d)$ , where d is the number of

entangling gate layers. Assume we made  $X_i$  measurements for calculating  $\langle h_i \rangle$ , its accuracy (standard deviation) would be  $\epsilon_i = 1/X_i$ . With  $X = \sum_{i=1}^{L} X_i$  measurements, the accuracy of Hamiltonian would be

$$\epsilon = \sum_{i=1}^{L} \frac{a_i}{\sqrt{X_i}} \le \sqrt{\sum_{i=1}^{L} a_i^2} \sqrt{\sum_{i=1}^{L} \frac{1}{X_i}} \le A \sqrt{\sum_{i=1}^{L} \frac{1}{X_i}}.$$
 (A28)

If  $X_i = X/L$ , we have  $\epsilon \le AL/\sqrt{X}$ , then we need  $X = A^2L^2/\epsilon^2 = A^2n^8/\epsilon^2$  measurements to achieve accuracy  $\epsilon$ . Considering the number of iterations for optimisation,  $N_{\text{iter}}$ , the total number of measurements is  $(A^2n^8/\epsilon^2)N_{\text{iter}}$ .