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Polariton spectra under the collective coupling regime. I. Efficient simulation of linear spectra and quantum dynamics



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ABSTRACT

We outline two general theoretical techniques to simulate polariton quantum dynamics and optical spectra under the collective coupling regimes described by a Holstein–Tavis–Cummings (HTC) model Hamiltonian. The first one takes advantage of sparsity of the HTC Hamiltonian, which allows one to reduce the cost of acting polariton Hamiltonian onto a state vector to the linear order of the number of states, instead of the quadratic order. The second one is applying the well-known Chebyshev series expansion approach for quantum dynamics propagation and to simulate the polariton dynamics in the HTC system; this approach allows us to use a much larger time step for propagation and only requires a few recursive operations of the polariton Hamiltonian acting on state vectors. These two theoretical approaches are general and can be applied to any trajectory-based non-adiabatic quantum dynamics methods. We apply these two techniques with our previously developed Lindblad-partially linearized density matrix approach to simulate the linear absorption spectra of the HTC model system, with both inhomogeneous site energy disorders and dipolar orientational disorders. Our numerical results agree well with the previous analytic and numerical work.

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I. INTRODUCTION

Coupling molecular excitations to a quantized radiation field inside an optical cavity produces a set of light–matter hybrid states known as polaritons. These polariton states, which are a hybridization of matter excitation and photonic excitation, have shown interesting photophysical properties. In particular, the light–matter interaction has been shown to effectively reduce the coupling between excitons and phonons, which is commonly referred to as the polaron decoupling effect.^{1,2} This polaron decoupling effect results in an enhanced charge transfer rate constant¹ and a reduction in the homogeneous linewidth of spectra^{3,4} and causes ballistic

exciton–polariton transport.^{5–7} In particular, the exciton–polariton coherence lifetime could be significantly prolonged due to reduced coupling with the phonon bath.^{4,8}

Linear spectroscopy^{9,10} and nonlinear spectroscopy^{4,11–15} are powerful measurements that provide a fundamental understanding of the photophysics of polariton systems. To capture the essential features of polariton photophysics, one needs to simulate N molecules collectively coupled to the cavity mode, typically using the Holstein–Tavis–Cummings (HTC) model Hamiltonian. Furthermore, one needs to explicitly incorporate exciton energy disorders (inhomogeneous disorder) and dipole orientational disorders for the molecule–cavity coupling interactions. In the typical

experimental setup, one couples $N = 10^3$ – 10^4 nanoplatelets to the cavity,^{8,16–19} or at least $N = 10^6$ organic molecules to one cavity mode. When there is no disorder present in the HTC Hamiltonian, one can in principle take advantage of the permutational symmetry of the problem and solve the problem with a mean field solution^{20,21} or the recently proposed CUT-E approach^{22,23} for the zero-temperature case and with a single high-frequency vibration mode per molecule. With the presence of various disorders (and no apparent symmetry in the system), it is computationally challenging to directly simulate the polariton dynamics when there are a large N of molecules coupled to the cavity, although there is progress in effectively describing static disorders as energy bins while taking advantage of the symmetry in HTC-type Hamiltonian.^{22,23} The tensor train decomposition method²⁴ in principle could handle HTC-type dynamics with disorders but remains computationally expensive due to the full quantum mechanical treatment for all degrees of freedom (DOFs).

In our previous work, we have combined the partial linearized density matrix (PLDM) approach^{25–28} with stochastic Lindblad dynamics to simulate polaritonic spectroscopy in lossy cavities. To include the cavity loss dynamics, we develop stochastic Lindblad dynamics, which exactly reproduce the Lindblad dynamics when averaged over an ensemble of trajectories. With simulated linear and 2D Electronic Spectra (2DES) of the polariton, we systematically investigated the influence of the light–matter coupling strength and cavity loss rate on the optical signals. In particular, we demonstrate that the polariton coherence (interpreted from the off-diagonal peak of the 2D electronic spectra) can be significantly prolonged by increasing the light–matter coupling strength, providing a theoretical verification of the polaron decoupling effect.⁴ We now aim to generalize our approach to the collective coupling situation, where a large N of molecules are collectively coupled to a cavity mode, with the presence of various types of molecular disorders.

In a series of two papers, we aim to develop an efficient and accurate approach to simulate the nonlinear spectra of the molecule–cavity hybrid system under the collective coupling regime. In Paper I (this work), we will focus on introducing two theoretical techniques that allow one to significantly reduce the computational cost of simulating the quantum dynamics of the HTC system with various types of disorders. We will focus on the linear spectra simulation. In Paper II, we will outline the theoretical details of stimulating 2DES for the polariton system under the collective coupling regime, with new computational techniques for the focusing algorithm of the PLDM simulation.

In this paper (Paper I), we outline two general theoretical techniques. The first one is related to the sparsity of the HTC Hamiltonian, which allows one to reduce the cost for the action of polariton Hamiltonian onto a state vector to the linear order in the number of states, instead of the quadratic order. The second one is applying the well-known Chebyshev series expansion approach for quantum dynamics propagation and to simulate the polariton dynamics in the HTC system. The Chebyshev expansion approach allows us to use a much larger time step for propagation and only requires a few recursive operations of the polariton Hamiltonian acting on state vectors, which can further take advantage of the first technique we developed in this work. These two theoretical approaches are general and can be applied to any trajectory-based non-adiabatic quantum dynamics methods^{29–34} or Gaussian wavepacket-based approaches.^{35,36} We

focus on simulating the linear absorption spectra of the HTC model system, with both inhomogeneous site-energy disorders and dipolar orientational disorders. With the significant reduction in the computational costs, we can directly simulate collective polariton dynamics with up to $N = 10^5$. Our numerical results agree well with the previous analytic³⁷ and numerical work.³⁸

II. POLARITON QUANTUM DYNAMICS

A. Model Hamiltonian

We describe the system of N -molecules coupled to a single cavity mode by the Holstein–Tavis–Cummings³⁹ Hamiltonian

$$\hat{H}_{\text{HTC}} = \hat{H}_s + \hat{H}_b + \hat{H}_{\text{sb}}, \quad (1)$$

where \hat{H}_s is the Hamiltonian of the system, \hat{H}_b is the Hamiltonian for bath DOF, and \hat{H}_{sb} is the system–bath interaction. The system Hamiltonian consists of excitonic and photonic DOFs,

$$\hat{H}_s = (\hat{H}_{\text{ex}} + \hat{H}_{\text{ph}}) \otimes \hat{\mathbb{1}}_b + \hat{H}_{\text{ex-ph}} \otimes \hat{\mathbb{1}}_b, \quad (2)$$

where \hat{H}_{ex} is the molecular (excitonic) Hamiltonian, \hat{H}_{ph} is the photonic Hamiltonian, $\hat{H}_{\text{ex-ph}}$ is the exciton–photon interaction, and $\hat{\mathbb{1}}_b$ is the identity operator in the bath subspace. The excitonic Hamiltonian consists of N uncoupled molecules, each with an onsite energy ϵ_n . The excitonic Hamiltonian can be written as

$$\hat{H}_{\text{ex}} = \sum_n \epsilon_n \hat{\sigma}_n^\dagger \hat{\sigma}_n \otimes \hat{\mathbb{1}}_{\text{ph}}. \quad (3)$$

Here, $\hat{\sigma}_n^\dagger = |e_n\rangle\langle g_n|$ and $\hat{\sigma}_n = |g_n\rangle\langle e_n|$ create and annihilate an excitation on the n th molecule, respectively, with $|g_n\rangle$ and $|e_n\rangle$ as the ground and excited states for molecule n , and $\hat{\mathbb{1}}_{\text{ph}}$ is the identity operator for the photonic subspace. The photonic Hamiltonian is described by a single cavity mode,

$$\hat{H}_{\text{ph}} = \hat{\mathbb{1}}_{\text{ex}} \otimes \hbar\omega_c \left(\hat{a}^\dagger \hat{a} + \frac{1}{2} \right), \quad (4)$$

where \hat{a}^\dagger and \hat{a} create and annihilate a photonic excitation (a photon) associated with the cavity mode, respectively, with energy $\hbar\omega_c$, and $\hat{\mathbb{1}}_{\text{ex}}$ is the identity operator for the excitonic subspace. The operators obey the commutation relation $[\hat{a}, \hat{a}^\dagger] = \hat{\mathbb{1}}_{\text{ph}}$. We further assume the rotating wave approximation for the light–matter coupling term,

$$\hat{H}_{\text{ex-ph}} = \sum_n \hbar g_c^n \left(\hat{\sigma}_n^\dagger \hat{a} + \hat{\sigma}_n \hat{a}^\dagger \right), \quad (5)$$

with g_c^n being the coupling strength for the n th exciton,

$$g_c^n = \sqrt{\frac{\hbar\omega_c}{2\mathcal{V}\epsilon_c}} \hat{\mu}_n \cdot \hat{\mathbf{e}} = \sqrt{\frac{\hbar\omega_c}{2\mathcal{V}\epsilon_c}} \mu_n \cdot \cos \theta_n. \quad (6)$$

Here, μ_n is the magnitude of the transition dipole moment of the n th exciton, $\hat{\mathbf{e}}$ is the field polarization direction, and θ_n is the angle between $\hat{\mu}_n$ and $\hat{\mathbf{e}}$. Furthermore, \mathcal{V} is the mode volume of the cavity and ϵ_c is the permittivity inside the cavity.

In addition, each molecular exciton is also coupled to a number of phonon modes that constitute the molecular bath. The bath Hamiltonian is described by

$$\hat{H}_b = \hat{1}_s \otimes \left[\frac{1}{2} \sum_{n=1}^N \sum_v (\hat{P}_{n,v}^2 + \omega_{n,v}^2 \hat{R}_{n,v}^2) \right], \quad (7)$$

where $\hat{P}_{n,v}$ and $\hat{R}_{n,v}$ are the momentum and position operators of the v th phonon on the n th matter site of frequency $\omega_{n,v}$ and $\hat{1}_s = \hat{1}_{\text{ex}} \otimes \hat{1}_{\text{ph}}$ is the identity operator of the “system” subspace that includes the excitonic and photonic DOFs. These bath DOFs are coupled bi-linearly to the excitonic sites providing diagonal fluctuations to on-site energies,

$$\hat{H}_{\text{sb}} = \sum_n \left(\hat{\sigma}_n^\dagger \hat{\sigma}_n \otimes \hat{1}_{\text{ph}} \right) \otimes \left(\sum_v C_{n,v} \hat{R}_{n,v} \right), \quad (8)$$

where $C_{n,v}$ is the strength of coupling between the n th exciton and the v th phonon (vibrational mode). For the n th exciton, the phonon frequencies and coupling strength are sampled from the Debye spectral density⁴⁰

$$J_n(\omega) = \frac{\pi}{2} \sum_v \frac{C_{n,v}^2}{\omega_{n,v}} \delta(\omega - \omega_{n,v}) = \frac{2\lambda_b \omega_b \omega}{\omega_b^2 + \omega^2}, \quad (9)$$

with λ_b being the bath reorganization energy and ω_b being the characteristic frequency of the bath. To perform linear and non-linear spectroscopy simulations, we use the matter dipole operator for computing multi-time correlation functions,^{41–45}

$$\hat{\mu} = \sum_n \mu_n \left(\hat{\sigma}_n^\dagger + \hat{\sigma}_n \right) \otimes \hat{1}_{\text{ph}}, \quad (10)$$

where μ_n is the magnitude of the transition dipole of excitation for the n th molecule. Cavity loss is not explicitly considered in the Hamiltonian. The loss effect is incorporated through the stochastic Lindblad approach developed in some previous work,^{46,47} with a brief summary provided in [Appendix A](#).

B. Polariton states

The diabatic eigenstates of \hat{H}_s are defined as

$$\hat{H}_s |\alpha\rangle = \epsilon_\alpha |\alpha\rangle. \quad (11)$$

When only considering the single excitation subspace $|j\rangle = \{|g\rangle \otimes |1\rangle, |e_n\rangle \otimes |0\rangle\}$ (where $|g\rangle \equiv \otimes |g_n\rangle$) with degenerate exciton energy $\epsilon_n = \epsilon$ and identical light–matter couplings $g_n^i = g_c$, Eq. (11) has a well-known analytic solution, with two bright polariton states^{10,48–51}

$$|+\rangle = \cos \Theta_N \left[\frac{1}{\sqrt{N}} \sum_{n=1}^N |e_n\rangle \otimes |0\rangle \right] + \sin \Theta_N |g\rangle \otimes |1\rangle, \quad (12a)$$

$$|-\rangle = -\sin \Theta_N \left[\frac{1}{\sqrt{N}} \sum_{n=1}^N |e_n\rangle \otimes |0\rangle \right] + \cos \Theta_N |g\rangle \otimes |1\rangle, \quad (12b)$$

where the mixing angle is

$$\Theta_N = \frac{1}{2} \tan^{-1} \left[\frac{2\sqrt{N}\hbar g_c}{\hbar\omega_c - \epsilon} \right] \in \left[0, \frac{\pi}{2} \right), \quad (13)$$

and $N - 1$ dark states for $k \in \{1, \dots, N - 1\}$,

$$|D_k\rangle = \frac{1}{\sqrt{N}} \sum_{n=1}^N \exp\left(-2\pi i \frac{nk}{N}\right) |e_n\rangle \otimes |0\rangle. \quad (14)$$

The polariton and dark states have the energy of

$$\epsilon_\pm = \frac{\epsilon + \hbar\omega_c}{2} \pm \frac{1}{2} \sqrt{(\epsilon - \hbar\omega_c)^2 + 4N\hbar^2 g_c^2}, \quad (15a)$$

$$\epsilon_{D_k} = \epsilon, \quad (15b)$$

where the eigenenergies of polariton states are split from the original exciton energy and the dark states have degenerate energies that are identical to the exciton energy. As shown in Ref. 49, the analytic solution^{10,48,49} is also available for \hat{H}_s . For the zero-detuning case $\epsilon = \hbar\omega_c$, the Rabi splitting is $\hbar\Omega_R = \epsilon_+ - \epsilon_- = 2\sqrt{N}\hbar g_c$, and the strong coupling limit^{39,50} is achieved when $\Omega_R \gg \frac{1}{2}(\Gamma + \kappa)$, where Γ is the cavity loss rate (linewidth of the cavity transmission spectra, $\tau_c = 1/\Gamma$ is the cavity lifetime) and κ is the exciton linewidth.

Note that the polariton states defined in Eq. (12) and the dark states defined in Eq. (14) are diabatic in nature, because there is no nuclear configuration dependence. In this case, the exciton–phonon couplings \hat{H}_{sb} will couple these polariton states and make transitions among them.^{8,51,52} Meanwhile, one can also choose to define $\hat{H}_s + \hat{H}_{\text{sb}}$ as the polariton Hamiltonian, whose eigenvector will be adiabatic polariton and dark states, and both of their characters will parametrically depend on the nuclear configuration.⁵³ In this case, the phonon fluctuation caused by \hat{H}_{sb} on polariton and dark states will be counted as the adiabatic polariton energy fluctuations, and the transitions among these adiabatic polariton and dark states are caused by the nuclear kinetic energy operators (as the derivative couplings). This is the picture used in the theoretical simulations in Ref. 53 (see Fig. 4 in that reference), where the “adiabatic dark state” is no longer purely dark, but can be called the “gray-dark states” because they have a little photonic contribution. Meanwhile, once considering static disorder in exciton energy, then one starts to lose the symmetry of the HTC Hamiltonian, and even the diabatic definition of the dark states will become non-dark (as can be observed in the linear spectra; see Fig. 6(b) for $\sigma = 0.2$ eV static disorder).

C. Trajectory based quantum dynamics approach

In this work, we consider the trajectory-based quantum dynamics simulations for the HTC model, which treat $\hat{H}_s + \hat{H}_{\text{sb}}$ as the quantum subsystem,

$$\hat{H}_Q = \hat{H}_s + \hat{H}_{\text{sb}}, \quad (16)$$

and \hat{H}_b as the classical subsystem. We want to explicitly consider the static energy disorders in site energy ϵ_n [inhomogeneous disorder, see Eq. (75)] as well as the dipole orientation disorder in $\cos \theta_n$ for the light–matter interaction terms.

In particular, we will use the partial linearized density matrix (PLDM) dynamics method,^{25–27,47,54,55} but the outlined theoretical schemes can be applied to any mixed quantum–classical (MQC) method, including the mean-field Ehrenfest dynamics, the trajectory surface hopping approach,^{56,57} or trajectory-based semiclassical methods such as the linearized semiclassical approach^{58–61} and symmetrical quasiclassical approach (SQC).^{62,63}

These trajectory-based quantum dynamics methods typically require the quantum propagation of the quantum state vector,

$$i\hbar \frac{\partial}{\partial t} |\Psi(t)\rangle = \hat{H}_Q(\mathbf{R}(t)) |\Psi(t)\rangle. \quad (17)$$

For a given diabatic basis $\{|j\rangle\}$ [for example, Eqs. (22)–(24)], one can expand $|\psi(t)\rangle = \sum_j c_j(t)|j\rangle$, and Eq. (17) becomes

$$i\hbar \dot{c}_k(t) = \sum_j \langle k | \hat{H}_Q(\mathbf{R}(t)) | j \rangle \cdot c_j(t). \quad (18)$$

Note that both \hat{H}_Q and $|\Psi(\mathbf{R}(t))\rangle$ depend parametrically on the bath configuration, $\mathbf{R}(t) \equiv \{R_{n,v}(t)\}$. The nuclear DOF is updated through the classical equations of motion, with different types of forces that depend on $\nabla_{\mathbf{R}}(\hat{H}_{sb} + \hat{H}_b)$, such as Ehrenfest dynamics (and similarly for other mean-field-like approaches) with $\mathcal{F}_{jk} = -c_j^* c_k \nabla_{\mathbf{R}}(j | (\hat{H}_{sb} + \hat{H}_b) | k)$, and for the HTC Hamiltonian special case, the nuclear force on the bath modes of the j th exciton is

$$\mathcal{F}_j = -|c_j|^2 \nabla_{\mathbf{R}}(j | (\hat{H}_{sb} + \hat{H}_b) | j), \quad (19)$$

which is purely diagonal due to the diagonal \hat{H}_{sb} in the diabatic basis [Eqs. (22)–(24)]. For the trajectory surface hopping approach, the nuclear force is $\mathcal{F} = -\nabla_{\mathbf{R}}(J(\mathbf{R}) | (\hat{H}_{sb} + \hat{H}_b) | J(\mathbf{R}))$, where $|J(\mathbf{R})\rangle$ is an active adiabatic surface (an eigenstate of \hat{H}_Q) that is determined stochastically through a given algorithm (such as the most widely used fewest switches algorithm^{56,57}).

The main challenge of solving Eq. (18) for the HTC Hamiltonian under the collective coupling regime is the large number of N that one has to incorporate. The right-hand side of the differential equation, $\hat{H}_Q|\Psi\rangle$, requires the operation of \hat{H}_Q matrix on a state vector $c_j(t)$. For the HTC Hamiltonian with N molecules and one cavity mode, there are $\mathcal{K} = N + 1 \propto \mathcal{O}(N)$ states in the single excitation subspace and $\mathcal{K} = N(N - 1)/2 + N + 1 \propto \mathcal{O}(N^2)$ states in the double excitation subspace. Meanwhile, the force evaluation for a mean-field-like approach of the HTC model [Eq. (19)] only scales with $\mathcal{O}(\mathcal{K})$ due to the purely diagonal structure of \hat{H}_{sb} , in the diabatic site basis. The typical experimental condition requires $N = 10^6 - 10^{12}$, resulting in a very large \mathcal{K} , and the computational work^{12,24,29,30,38} is often limited by $N = 10^1 - 10^2$ due to the computational cost.

The other possibility is to consider the propagator approach instead of solving the Time Dependent Schrodinger equation (TDSE) [Eq. (18)] directly. Note that \hat{H}_Q contains a static part of \hat{H}_s (with a very special symmetry) and a purely diagonal part \hat{H}_{sb} that depends on $\mathbf{R}(t)$. One can consider the following symmetrical Trotter splitting:

$$|\Psi(t + \Delta t)\rangle \approx e^{-\frac{i}{\hbar} \hat{H}_{sb} \Delta t / 2} e^{-\frac{i}{\hbar} \hat{H}_s \Delta t} e^{-\frac{i}{\hbar} \hat{H}_{sb} \Delta t / 2} |\Psi(t)\rangle, \quad (20)$$

with the accuracy up to $\mathcal{O}(\Delta t^3)$ [which can be easily shown by using the Baker–Campbell–Hausdorff formula in Eq. (11.197) from Ref. 64]. Using a diabatic basis $|j\rangle$ and the polariton basis $|\alpha\rangle$ [see Eq. (11)], one can rewrite Eq. (20) as

$$\begin{aligned} |\Psi(t + \Delta t)\rangle &= \sum_k c_k(t + \Delta t) |k\rangle \\ &\approx \sum_k |k\rangle \sum_{j\alpha} e^{-\frac{i}{\hbar} \lambda_k \Delta t / 2} \langle k | \alpha \rangle e^{-\frac{i}{\hbar} \epsilon_\alpha \Delta t} \langle \alpha | j \rangle e^{-\frac{i}{\hbar} \lambda_j \Delta t / 2} c_j(t), \end{aligned} \quad (21)$$

where ϵ_α is the diabatic polariton energy defined in Eq. (11), $\lambda_j = \langle j | \hat{H}_{sb}(\mathbf{R}) | j \rangle$ is the pure diagonal contribution of the system–bath coupling (that parametrically depends on \mathbf{R}), and $c_j(t) = \langle j | \Psi(t) \rangle$. In principle, one needs to diagonalize \hat{H}_s to obtain ϵ_α and $\{\langle \alpha | j \rangle\}$ at a cost of $\mathcal{O}(\mathcal{K}^2)$ [see Refs. 65–67, due to the symmetry of \hat{H}_s , instead of the usual cost of $\mathcal{O}(\mathcal{K}^3)$ for diagonalizing an arbitrary $\mathcal{K} \times \mathcal{K}$ matrix]. Nevertheless, because these are diabatic quantities, one can in principle compute them only once and store them. Of course, when only single- and double-excitation subspaces are considered, these results are (almost) analytically available.^{10,48,49} When only considering the single excitation subspace $|j\rangle = \{|g\rangle \otimes |1\rangle, |e_n\rangle \otimes |0\rangle\}$, the expression of $S_{kj}(\Delta t) \equiv \sum_\alpha \langle k | \alpha \rangle e^{-\frac{i}{\hbar} \epsilon_\alpha \Delta t} \langle \alpha | j \rangle$ is analytically available and can be precomputed and stored. One just needs to multiply the $\tilde{S}_{kj} \equiv e^{-\frac{i}{\hbar} \lambda_k \Delta t / 2} S_{kj}(\Delta t) e^{-\frac{i}{\hbar} \lambda_j \Delta t / 2}$ matrix onto the $c_j(t)$ column matrix to get the result of $c_k(t + \Delta t)$, which in principle still requires a cost of $\mathcal{O}(N^2)$.

As such, for the quantum dynamics problem of the HTC system, the main computational bottleneck is the $\hat{H}_Q|\Psi\rangle$ evaluation. One would also want to avoid directly diagonalizing \hat{H}_Q , as well as increase the time step Δt for the dynamics propagation as much as possible (because each step requires at least one $\hat{H}_Q|\Psi\rangle$ evaluation) while keeping the propagation stable. In Sec. III, we introduce several techniques that can significantly reduce the computational costs of propagating Eq. (17). These techniques can be broadly applied to any MQC trajectory-based methods for polariton quantum dynamics simulations.

III. THEORETICAL APPROACHES

We demonstrate strategies to efficiently manipulate matrix–vector multiplications to reduce the computation cost of semiclassical simulations. When considering the single excitation subspace, the size of the Hilbert space scales as $\mathcal{K} \propto \mathcal{O}(N)$, and solving the TDSE in principle requires the operation of $\mathcal{O}(\mathcal{K}^2) \propto \mathcal{O}(N^2)$. When considering the Hilbert space up to the second excitation subspace (for 2DES simulations in Paper II) $\mathcal{K} \propto \mathcal{O}(N^2)$, the quantum dynamics simulation will in principle require a cost of $\mathcal{O}(\mathcal{K}^2) \propto \mathcal{O}(N^4)$. To reduce the scaling of the cost, we first demonstrate how the action of \hat{H}_Q on $|\Psi(t)\rangle$ can be written as the sum of a simple Hadamard product of vectors that reduces matrix–vector multiplication from $\mathcal{O}(N^2)$ to $\mathcal{O}(N)$ for single excitation subspace and $\mathcal{O}(N^4)$ to $\mathcal{O}(N^2)$ for up to double excitation subspace. We demonstrate a similar scaling reduction for the action of the dipole operator on the wavefunction, which is relevant for linear and non-linear spectra calculations. We further apply a Chebyshev series expansion approach to simulate the polariton dynamics in the HTC system. The Chebyshev expansion approach allows us to use a much larger time step for propagation and only requires a few recursive operations of the polariton Hamiltonian acting on state vectors, which can further take advantage of the scaling reduction due to the sparsity of the HTC Hamiltonian.

The theoretical approach we outlined here can be widely applicable to any mixed quantum–classical methods to simulate polariton dynamics (population, coherence, and different types of correlation functions). The scaling reduction in the single excitation subspace is the most relevant part of the linear spectra simulation in this paper,

whereas the scaling reduction in the double excitation subspace will be relevant for 2DES simulations in Paper II.

A. Efficient evaluation of $\hat{H}_Q|\Psi\rangle$ in the double excitation subspace

The quantum dynamics propagation requires a basic operator action of $\hat{H}_Q|\Psi\rangle$ operation [see Eqs. (17) and (18)] using, for example, the RK4 method, or $\hat{O}|\Psi\rangle$ [see Eq. (36)] in the Chebyshev expansion scheme [Eq. (35)]. For a \mathcal{K} -dimensional system in \hat{H}_Q , this would in principle require \mathcal{K}^2 operations in matrix multiplications. However, we notice that the matrix of \hat{H}_Q in the HTC Hamiltonian is extremely sparse in the single and double excitation subspaces, which are the relevant subspaces for photophysics and spectroscopy measurements. Here, we take advantage of this sparsity to significantly reduce the scaling of the computational costs of quantum dynamics propagation.

We take the bare excitonic states and the photonic Fock states to form the diabatic basis. The collective ground state can be written as

$$|G^0\rangle = \left(\bigotimes_n |g_n\rangle \right) \otimes |0\rangle. \quad (22)$$

Within the single excitation manifold, we can describe two kinds of diabatic states characterized by either purely matter or purely photonic excitation character,

$$|E_n^0\rangle = \left(\bigotimes_{m \neq n} |g_m\rangle \right) \otimes |e_n\rangle \otimes |0\rangle, \quad (23a)$$

$$|G^1\rangle = \left(\bigotimes_n |g_n\rangle \right) \otimes |1\rangle. \quad (23b)$$

In the second excitation manifold, we get three additional types of states, with either double matter excitations, double photon excitation, or a mixed matter–photon excited state, expressed as follows:

$$|E_{nm}^0\rangle = \left(\bigotimes_{p \neq n,m} |g_p\rangle \right) \otimes |e_n\rangle \otimes |e_m\rangle \otimes |0\rangle, \quad (24a)$$

$$|E_n^1\rangle = \left(\bigotimes_{m \neq n} |g_m\rangle \right) \otimes |e_n\rangle \otimes |1\rangle, \quad (24b)$$

$$|G^2\rangle = \left(\bigotimes_n |g_n\rangle \right) \otimes |2\rangle. \quad (24c)$$

With the above-mentioned basis, a general quantum state can be expanded in this double-excitation subspace as

$$|\Psi\rangle = c_{G^0}|G^0\rangle + \sum_n c_{E_n^0}|E_n^0\rangle + c_{G^1}|G^1\rangle + \sum_{n,m>n} c_{E_{nm}^0}|E_{nm}^0\rangle + \sum_n c_{E_n^1}|E_n^1\rangle + c_{G^2}|G^2\rangle, \quad (25)$$

where $c_j = \langle j|\Psi\rangle$. From now on, we will follow the order of the basis set expansion outlined in Eq. (25) and represent $|\Psi\rangle$ as

$$|\Psi\rangle \rightarrow \begin{bmatrix} c_{G^0} \\ c_{E_n^0} \\ c_{G^1} \\ c_{E_{nm}^0} \\ c_{E_n^1} \\ c_{G^2} \end{bmatrix}. \quad (26)$$

The quantum Hamiltonian \hat{H}_Q in Eq. (50), when represented in the double excited diabatic basis, is very sparse in nature. Although the size of the Hamiltonian increases as $\mathcal{O}(N^4)$ for N molecules, the sparsity of the Hamiltonian (defined as the ratio between the number of nonzero matrix elements and the total number of matrix elements) scales as $1 - 1/N^2$ and very quickly converges to 1.

Figure 1 demonstrates the increase in sparsity as N increases, with the nonzero matrix elements explicitly shown in a darker blue color, and with the basis order explicitly following Eq. (25). We can precompute the action of the sparse Hamiltonian, Eq. (50), on a general system quantum state in Eq. (25),

$$\begin{aligned} \hat{H}_Q|\Psi\rangle = & \sum_n (c_{E_n^0} \cdot \epsilon_n + \hbar g_c^n c_{G^1}) |E_n^0\rangle \\ & + \left(\hbar \omega_c \cdot c_{G^1} + \sum_n g_c^n c_{E_n^0} \right) |G^1\rangle \\ & + \sum_{n,m>n} [(\epsilon_n + \epsilon_m) \cdot c_{E_{nm}^0} + \hbar (g_c^n c_{E_n^0} + g_c^m c_{E_m^0})] |E_{nm}^0\rangle \\ & + \sum_n \left\{ \sum_{m>n} \hbar g_c^m c_{E_{nm}^0} + (\epsilon_n + \hbar \omega_c) \cdot c_{E_n^1} + \sqrt{2} \hbar g_c^n c_{G^2} \right\} |E_n^1\rangle \\ & + \left(2 \hbar \omega_c \cdot c_{G^2} + \sqrt{2} \hbar \sum_n g_c^n c_{E_n^1} \right) |G^2\rangle, \quad (27) \end{aligned}$$

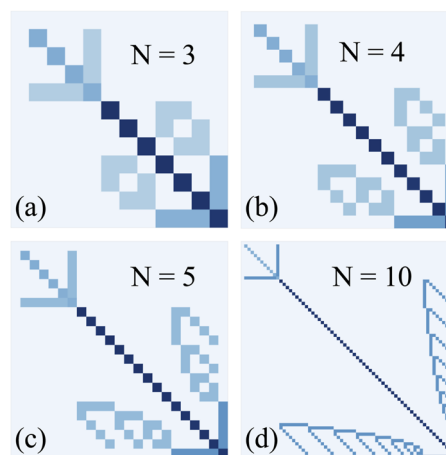


FIG. 1. HTC Hamiltonian structure for N molecules. The light cyan color represents the zero matrix elements, whereas the darker blue represents the non-zero matrix elements. The order of the basis is the same as indicated in Eq. (25): from left to right for the row and from top to bottom for the column. Panels (a)–(d) represent the Hamiltonian for $N = 3, 4, 5,$ and 10 molecules, respectively, within the double excitation subspace.

where

$$\epsilon_n = \epsilon_n + \sum_v C_{n,v} R_{n,v}$$

is the exciton energy ϵ_n plus the bath fluctuation, with the instantaneous value of the bath coordinate $R_{n,v}$ (that is fixed during the Δt propagation). The expansion in Eq. (27) can be expressed as the following Hadamard product (represented by the symbol \odot):

$$\hat{H}_Q|\Psi\rangle = \underbrace{|\epsilon\Psi\rangle}_{\text{diagonal}} \odot |\Psi\rangle + \underbrace{\hbar g_c|\Phi\rangle}_{\text{off-diagonal}}, \quad (28)$$

which is the *first key result* of this work. In Eq. (28), the diagonal energies are accounted for by the vector

$$|\epsilon\Psi\rangle \rightarrow \begin{bmatrix} 0 \\ \epsilon_{E_n^0} \\ \hbar\omega_c \\ \epsilon_{E_{nm}^0} \\ \epsilon_{E_n^1} \\ 2\hbar\omega_c \end{bmatrix}. \quad (29)$$

Here, $\epsilon_{E_n^0} = [\epsilon_1, \epsilon_2, \dots, \epsilon_N]^T$ is the vector of exciton energies, $\epsilon_{E_n^0} = [\epsilon_{12}, \epsilon_{13}, \dots, \epsilon_{(N-1)N}]^T$ is the vector of double exciton energies, and $\epsilon_{E_n^1} = [\epsilon_1 + \omega_c, \epsilon_2 + \omega_c, \dots, \epsilon_N + \omega_c]^T$ is the vector of collective matter and photon excited state energies.

Furthermore, the off-diagonal action of \hat{H}_Q on $|\Psi\rangle$ is accounted by the vector

$$|\Phi\rangle \rightarrow \begin{bmatrix} 0 \\ c_{G^1} \cdot \mu_n \cos \theta_n \\ \sum_n c_{E_n^0} \cdot \mu_n \cos \theta_n \\ c_{E_n^1} \cdot \mu_n \cos \theta_n + c_{E_m^1} \cdot \mu_m \cos \theta_m \\ \sum_{m>n} c_{E_{nm}^0} \cdot \mu_m \cos \theta_m + \sqrt{2}c_{G^2} \cdot \mu_n \cos \theta_n \\ \sqrt{2}\sum_n c_{E_n^1} \cdot \mu_n \cos \theta_n \end{bmatrix}, \quad (30)$$

and when there is no dipole orientation disorders, simply set $\cos \theta_n = 1$ in the above expression.

The operation in Eq. (28) has a *linear* scaling in the number of states $\mathcal{O}(K)$ and thus $\propto \mathcal{O}(N)$ for a single excitation space and $\propto \mathcal{O}(N^2)$ for a double excitation space when having N -molecules for the polaritonic systems. The action of \hat{H}_Q on $|\Psi\rangle$ as outlined in Eq. (28) can be used to accurately calculate electronic evolution via a simple RK4 solution of Eq. (17) or through polynomial expansions of exponential propagators such as the Chebyshev series, which will be discussed next.

For realistic molecules in the collective coupling regime, intermolecular interactions need to be considered. These intermolecular interactions will make the off-diagonal elements in Fig. 1 non-zero. Nevertheless, in most of the recent experiments,⁶ the density of the chromophores is very low, and the intermolecular distance is estimated to be at least 40 Å; thus, even the long-range Förster type of dipole-dipole interaction is negligible. Even when explicitly considering these Förster-type dipole-dipole couplings, they can be effectively described by introducing simple off-diagonal coupling

terms in the HTC Hamiltonian, which only couples a molecule with a few of its nearest neighbors. As such, the N -molecule HTC Hamiltonian is still sparse with only a few band-diagonal coupling and it can again be recast with some additional simple vector-vector Hadamard products, so we can utilize this sparsity. This will be explored in the future.

Furthermore, we want to clarify that for linear spectra calculations in this paper, the only important states are the states with energies similar to that of one exciton, and one does not need to include the excited states in the second excitation manifold. We develop the general framework that includes the double excitation subspace for the future non-linear spectra calculation, which is key to an accurate description of the excited state absorption (ESA) calculation for 2DES. This will be discussed in our forthcoming paper.

B. Chebyshev series expansion for quantum dynamics propagation

To solve TDSE in Eq. (17), one can use many numerical methods, such as the RK4 solution of Eq. (17) or using the symplectic integrator.⁶⁸ These approaches can still take advantage of the sparsity properties of the Hamiltonian outlined in Sec. III A. The numerical challenges for these propagation-based methods are the requirement of relatively small Δt for stable propagation (see Appendix B for benchmark). Due to the large N considered in the HTC Hamiltonian, each propagation step remains computationally expensive.

To use a relatively large Δt , we use the Chebyshev polynomial expansion⁶⁹ approach to solve the TDSE. During a time Δt where the nuclear configuration \mathbf{R} is fixed, the propagation can be expressed as

$$|\Psi(t + \Delta t)\rangle = \hat{U}(\Delta t)|\Psi(t)\rangle = \exp\left(-\frac{i}{\hbar}\hat{H}_Q\Delta t\right)|\Psi(t)\rangle, \quad (31)$$

where the Chebyshev expansion of the evolution operator is expressed⁶⁹ as

$$\hat{U}(\Delta t)|\Psi\rangle = b_0(z)|\Psi^{(0)}\rangle + \sum_{n=1}^{\infty} \phi_n(z)b_n(z)|\Psi^{(n)}\rangle, \quad (32)$$

where $|\Psi^{(0)}\rangle = |\Psi(t)\rangle$ and $\{b_n\}$ are the Bessel-function coefficients of the first type. The number of coefficients required for the above expansion is dictated by the (energy-time) uncertainty parameter

$$z = \frac{\delta_E \cdot \Delta t}{2\hbar}, \quad (33)$$

where δ_E is the spectral radius (the energy difference between the highest and the lowest eigenenergies for \hat{H}_Q). For $z \leq 1$, one only needs to include the first few terms of the expansion. In the Chebyshev expansion approach, a large time step Δt or a higher spectral radius δ_E results in a higher z value, which necessitates the requirement of more number of terms in the expansion, Eq. (32).

Furthermore, $\phi_n(z)$ in Eq. (32) is a phase factor that contains information on the shift in the energy axis due to the normalization of the Hamiltonian, given by

$$\phi_n(z) = 2i^n \exp\left(\frac{i}{\hbar}\left[\frac{\delta_E}{2} + \epsilon_{\min}\right]\Delta t\right). \quad (34)$$

Here, δ_E corresponds to the approximate spectral radius of \hat{H}_Q and ϵ_{\min} is an estimate of the lowest eigenvalue of \hat{H}_Q . For the Chebyshev method, one only needs a rough estimate for δ_E and ϵ_{\min} , which requires a rough estimation of the lowest and highest eigenvalues of \hat{H}_Q .

The n th order term of the Chebyshev expansion in Eq. (32) is calculated by the recursion relation,⁶⁹

$$|\Psi^{(n)}\rangle = \begin{cases} \hat{O}|\Psi\rangle, & n = 1, \\ 2\hat{O}|\Psi^{(n-1)}\rangle - |\Psi^{(n-2)}\rangle, & n \geq 2, \end{cases} \quad (35)$$

where \hat{O} is the modified (normalized) Hamiltonian,

$$\hat{O} = 2 \frac{\hat{H}_Q - \epsilon_{\min} \hat{1}_s}{\delta_E} - \hat{1}_s. \quad (36)$$

The action of \hat{O} on $|\Psi\rangle$ is the same as \hat{H}_Q on $|\Psi\rangle$ as described in Eq. (28), with the coefficients now scaled according to Eq. (36).

Figure 2 presents the magnitude of the Chebyshev coefficients as a function of the expansion term index for various z -values. With an increase in z , more terms are needed for the expansion to converge. For $z \ll 1$, only the first three or four terms in the expansion are required. For comparison, a typical propagation scheme such as RK4 or a symplectic integrator (velocity Verlet) typically needs a much smaller substep, $\delta t \approx \Delta t/10 - \Delta t/100$, for a stable integration of the dynamics. The Chebyshev expansion approach thus provides a significant speedup of the dynamics propagation compared to solving Eq. (31) using RK4 because one can use a much larger Δt (up to 100 times), with only a slight increase in the prefactor of the computational cost associated with how many terms are needed to form a converged Chebyshev expansion (each associated with the $\hat{O}|\Psi\rangle$ operations). In all simulations, we are in the regime of $z \ll 1$ and we found that only $n = 4$ terms of the Chebyshev polynomial are required to converge the results.

C. Decomposing single and double excitation subspaces

The inclusion of a double excitation subspace can make our δ_E very large, forcing us to choose a very small Δt . Note that the

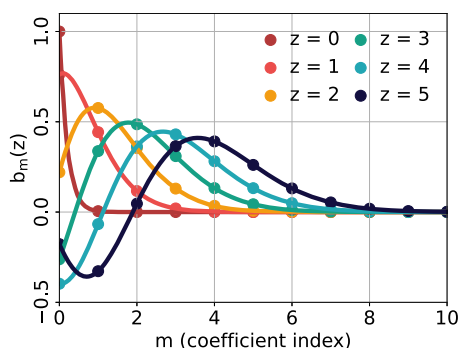


FIG. 2. Chebyshev coefficients as a function of expansion index for various uncertainty parameters, z . The colored curves represent the Bessel function for a continuous “ m ” coordinate, and the dots are the Bessel functions evaluated at integer “ m ” values.

double-excitation subspace has a diagonal energy that is about twice that of the single-excitation subspace (when it has zero light–matter detuning). Furthermore, the Hamiltonian has a block-diagonal structure of \hat{H}_Q , illustrated in Fig. 1, where the single and double excitation manifolds are not coupled to each other through Hamiltonian operators (they are connected through the dipole operator and will have transitions upon laser excitation). This observation leads us to separate the first and second excitation subspaces with their individual spectral radius. The operation $\hat{O}|\Psi\rangle$, via Eq. (28), is thus performed separately for the different subspaces. We separate the system Hilbert space into different excitation manifolds,

$$\hat{H}_Q = \bigoplus_{\xi} \hat{H}^{(\xi)}, \quad (37a)$$

$$|\Psi\rangle = \bigoplus_{\xi} |\Psi^{(\xi)}\rangle, \quad (37b)$$

where $\xi = 0$ indicates the ground state [Eq. (22)], and $\xi = 1$ and $\xi = 2$ indicate the single [Eq. (23)] and double excitation manifolds [Eq. (24)], respectively. The quantum propagation can now be separated into the propagation of different subspaces,

$$\hat{U}(\Delta t)|\Psi\rangle = \bigoplus_{\xi} \exp\left(\frac{-i\hat{H}^{(\xi)}\Delta t}{\hbar}\right)|\Psi^{(\xi)}\rangle. \quad (38)$$

We can now independently apply the Chebyshev expansion in different subspaces. For the first excitation subspace, the approximate spectral radius and lower-bound of the eigenvalue are given by

$$\delta_E^{(1)} = 2g_c\sqrt{N} + 2\lambda_b, \quad (39a)$$

$$\epsilon_{\min}^{(1)} = \bar{\epsilon} - g_c\sqrt{N} - \lambda_b, \quad (39b)$$

where $\bar{\epsilon} = \sum_n \epsilon_n/N$ and λ_b is the bath reorganization energy that accounts for the bath fluctuations. For the second excitation manifold, these quantities are^{10,48,49}

$$\delta_E^{(2)} = 2g_c\sqrt{2(N-1)} + 4\lambda_b, \quad (40a)$$

$$\epsilon_{\min}^{(2)} = 2\bar{\epsilon} - g_c\sqrt{2(N-1)} - 2\lambda_b, \quad (40b)$$

where we used the analytic results^{10,48,49} to estimate the lower bound of the polariton eigenstate as $2\bar{\epsilon} - g_c\sqrt{2(N-1)}$ in Eq. (40b).

The Chebyshev expansion allows efficient dynamics propagation by accurately calculating the action of the unitary propagator on an arbitrary vector. However, there is still a numerical challenge in simulating quantum dynamics with the HTC model. In Eq. (34), the phase factor in the Chebyshev expansion, Eq. (32), depends on the approximate absolute magnitude of the lower eigenvalue ϵ_{\min} . Thus, if the eigenvalue has a large absolute magnitude, the accuracy of this method is reduced due to the noise that builds up from the highly oscillating term coming from the high absolute magnitude of ϵ_{\min} , especially if we are simulating both ground state and single excitations (or further including the double excitation) where these

values are on the order of eV, and the difference between $\epsilon_{\min}^{(1)}$ and $\epsilon_{\min}^{(2)}$ is also on the order of eV. The same challenge also exists in other propagation methods, such as using RK4 to solve Eq. (17), because when a large energy difference exists in a Hamiltonian, one would need a much smaller Δt to fully capture the phase oscillations when propagating the coefficients.

This numerical challenge can be easily resolved by decomposing the Hamiltonian as $\hat{H} = \bar{\epsilon}\hat{1} + \Delta\hat{H}$, where $\bar{\epsilon}$ is a constant reference energy level and $\Delta\hat{H}$ is the fluctuation around $\bar{\epsilon}$. Because $\bar{\epsilon}\hat{1}$ commutes with $\Delta\hat{H}$, the quantum evolution becomes $e^{-\frac{i}{\hbar}\hat{H}\Delta t} = e^{-\frac{i}{\hbar}\bar{\epsilon}\Delta t} \cdot e^{-\frac{i}{\hbar}\Delta\hat{H}\Delta t}$, where the highly oscillatory part of the phase $e^{-\frac{i}{\hbar}\bar{\epsilon}\Delta t}$ will be evaluated analytically, and the rest of the propagation $e^{-\frac{i}{\hbar}\Delta\hat{H}\Delta t}$ will be numerically updated using the Chebyshev expansion. Furthermore, in the HTC Hamiltonian, these subspaces are not directly connected by Hamiltonian terms. Following this logic, we rewrite the Hamiltonian in Eq. (37a) as

$$\hat{H}_Q = \hat{H}^{(0)} \oplus (\bar{\epsilon}\hat{1}_s^{(1)} + \Delta\hat{H}^{(1)}) \oplus (2\bar{\epsilon}\hat{1}_s^{(2)} + \Delta\hat{H}^{(2)}), \quad (41)$$

where $\hat{1}_s^{(\xi)}$ is the identity operator of the ξ th excitation subspace and $\Delta\hat{H}^{(\xi)} = \hat{H}^{(\xi)} - (\xi\bar{\epsilon}) \cdot \hat{1}_s^{(\xi)}$, in which $\xi\bar{\epsilon}$ is the reference energy for the ξ th excitation subspace (i.e., $\bar{\epsilon}$ for single excitation and $2\bar{\epsilon}$ for double excitation subspace). Since the identity operator commutes with any other operators in its respective subspace, using Eq. (41), the propagator now becomes

$$\hat{U}(\Delta t)|\Psi\rangle = \bigoplus_{\xi} \exp\left(\frac{-i\xi\bar{\epsilon}\Delta t}{\hbar}\right) \exp\left(\frac{-i\Delta\hat{H}^{(\xi)} \cdot \Delta t}{\hbar}\right)|\Psi^{(\xi)}\rangle, \quad (42)$$

where the $\exp\left(\frac{-i}{\hbar}\Delta\hat{H}^{(\xi)}\Delta t\right)|\Psi^{(\xi)}\rangle$ update will be performed using the Chebyshev expansion and is then multiplied with a phase $\exp\left(\frac{-i}{\hbar}\xi\bar{\epsilon}\Delta t\right)$. With this separation, the accuracy of the Chebyshev propagator is independent of the absolute value of electronic excitation energies. For the Chebyshev expansion with the shifted eigenvalues, the approximate lower eigenvalues used in Eq. (36) are now

$$\epsilon_{\min}^{(1)} = -g_c\sqrt{N} - \lambda_b, \quad (43)$$

$$\epsilon_{\min}^{(2)} = -g_c\sqrt{2(N-1)} - 2\lambda_b, \quad (44)$$

whereas $\delta_E^{(\xi)}$ [as expressed in Eqs. (39a) and (40a)] remains unchanged. Because different subspaces are not directly coupled by \hat{H}_Q , when propagating the quantum dynamics governed by \hat{H}_Q , the above Chebyshev scheme will be used independently for each subspace. Meanwhile, different subspaces are connected through the cavity loss process (such as $|G^1\rangle \rightarrow |G^0\rangle$ in the single excitation subspace and $|G^2\rangle \rightarrow |G^1\rangle$ and $|E_n^1\rangle \rightarrow |E_n^0\rangle$ in the double excitation space), and when computing the response function, the instantaneous action of the dipole operator will connect them. For those two types of operations, one will apply the stochastic Lindblad algorithm (Appendix A) and the rule of dipole on state vectors directly, without shifting the reference energies. This simple technique outlined in Eq. (42) significantly reduces the noise of the numerical propagation

and increases the accuracy of the Chebyshev expansion method, as shown in Appendix B.

D. Spectroscopy simulations and efficient evaluation of the $\hat{\mu}|\Psi\rangle$ operation

Within the linear response limit, the absorption spectra of a system can be expressed as the dipole-dipole auto-correlation function^{25,41,44,47}

$$R^{(1)}(t_1) = i \text{Tr} [\hat{\mu}(t_1)[\hat{\mu}(0), \hat{\rho}_0]], \quad (45)$$

where the laser excitation is applied at time $t = 0$, and one detects the system response at time t_1 . The trace is performed for all DOFs. In addition, $\hat{\rho}_0$ represents the initial density matrix of the entire system in equilibrium ground state at $t = -\infty$, ($\hat{\rho}_0 = \hat{\rho}(-\infty)$). Furthermore, $\hat{\rho}_0$ is assumed to be a separable state between the polariton and the phonon bath DOFs as follows:

$$\hat{\rho}_0 = \hat{\rho}_g \otimes \hat{\rho}_b = |G^0\rangle\langle G^0| \otimes \hat{\rho}_b = |G^0\rangle\langle G^0| \otimes \frac{e^{-\beta\hat{H}_b}}{\mathcal{Z}}, \quad (46)$$

with $|G^0\rangle$ defined in Eq. (22).

The frequency domain spectra can be computed by

$$R^{(1)}(\omega) = \int_0^T dt e^{i\omega t} R^{(1)}(t) \cos\left(\frac{\pi t}{2T}\right), \quad (47)$$

where T is the maximum time of the recorded time-domain spectra and $\cos(\pi t/2T)$ is a smoothing function.^{25,47}

As can be seen, to compute $R^{(1)}(t_1)$ and $R^{(1)}(\omega)$, one needs to act the dipole operator, $\hat{\mu}$, on the density matrix, which in principle also scales as $\mathcal{O}(K^4)$. However, $\hat{\mu}$ also takes the form of a very sparse matrix in the single and double excitation subspaces, as illustrated in Fig. 3. The sparsity of the dipole operator scales as $1 - 1/N^2$ with N molecules coupled to a cavity mode. The action of $\hat{\mu}$ on a ket vector [following the same arrangement of the basis as in Eq. (25)] can be written as

$$\begin{aligned} \hat{\mu}|\Psi\rangle &= \sum_n \mu_n c_{E_n^0} |G^0\rangle + \sum_n \left(\mu_n c_{G^0} + \sum_{m>n} \mu_m c_{E_{nm}^0} \right) |E_n^0\rangle \\ &+ \left(\sum_n \mu_n c_{E_n^1} \right) |G^1\rangle + c_{G^1} \sum_n \mu_n |E_n^1\rangle. \end{aligned} \quad (48)$$

The above becomes a very simplified modification of $|\Psi\rangle$,

$$\hat{\mu}|\Psi\rangle = \begin{bmatrix} \sum_n \mu_n c_{E_n^0} \\ \mu_n c_{G^0} + \sum_{m>n} \mu_m c_{E_{nm}^0} \\ \sum_n \mu_n c_{E_n^1} \\ \sum_{m>n} \mu_m c_{E_{nm}^0} \\ \mu_n c_{G^1} \\ 0 \end{bmatrix}. \quad (49)$$

This helps us significantly reduce the cost of acting dipole operator on state vectors in spectroscopy calculations.

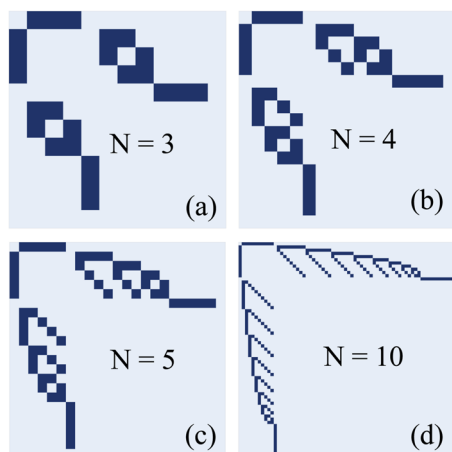


FIG. 3. Matrix representation of the dipole operator $\hat{\mu}$ [see Eq. (48)] for N molecules. The light cyan color represents the zero matrix elements, whereas the dark blue represents the non-zero matrix elements. Panels (a)–(d) show the Hamiltonian for 3, 4, 5, and 10 molecules, respectively, under the double excitation basis.

IV. COMPUTATIONAL METHOD

A. PLDM approach for dynamics propagation

We briefly outline the PLDM approach for quantum dynamics simulations.^{26,27,55,70} The diabatic Hamiltonian of a system coupled to a bath (nuclear DOF) can be expressed as

$$\begin{aligned}\hat{H} &= \frac{\hat{p}^2}{2M} + V_0(\hat{R}) + \sum_a^{\mathcal{K}} V_{aa}(\hat{R})|a\rangle\langle a| + \frac{1}{2} \sum_{b \neq a}^{\mathcal{K}} V_{ab}(\hat{R})|a\rangle\langle b| \\ &= \hat{H}_b + \hat{H}_Q,\end{aligned}\quad (50)$$

where \hat{R} and \hat{P} are the position and momenta, respectively, of the bath particles of mass M and $V_0(\hat{R})$ is the state-independent part of the Hamiltonian. In the context of this work,

$$\begin{aligned}\hat{H}_b &= \frac{\hat{p}^2}{2M} + V_0(\hat{R}), \\ \hat{H}_Q &= \sum_a^{\mathcal{K}} V_{aa}(\hat{R})|a\rangle\langle a| + \frac{1}{2} \sum_{b \neq a}^{\mathcal{K}} V_{ab}(\hat{R})|a\rangle\langle b|,\end{aligned}\quad (51)$$

where \hat{H}_Q is the quantum part of the Hamiltonian, which includes \hat{H}_s and \hat{H}_{sb} [Eq. (16)].

Using the Meyer–Miller–Stock–Thoss (MMST) mapping procedure,^{58,71,72} we get the *classical* mapping Hamiltonian,²⁶

$$\begin{aligned}\mathcal{H}(R, \mathbf{x}, \mathbf{p}) &= \frac{P^2}{2M} + V_0(R) + \frac{1}{2} \sum_a V_{aa}(R)(x_a^2 + p_a^2) \\ &\quad + \frac{1}{2} \sum_{b \neq a} V_{ab}(R)(x_a x_b + p_a p_b).\end{aligned}\quad (52)$$

This Hamiltonian, in principle, will give exact quantum dynamics when decoupled from the nuclear DOF and can be rigorously derived using the MMST mapping procedure.^{71,72} When applying the semiclassical approximation in PLDM,²⁶ a pre-factor of the

propagator exactly cancels the zero-point energy correction factor in the MMST mapping Hamiltonian, leading to Eq. (52) as the effective Hamiltonian in PLDM. PLDM has been shown to accurately capture non-adiabatic quantum dynamics in model systems^{26,27,55} and *ab initio* systems,⁷³ providing accurate charge transfer rate constant⁵⁴ and excitation energy transfer dynamics⁷⁰ and computing linear and non-linear spectroscopy.²⁵

For a given set of system operators \hat{A} and \hat{B} , and $\hat{\rho}_b$ being the initial bath density operator, the time evolution of the PLDM is

$$\begin{aligned}C_{AB}(t) &= \text{Tr} \left[\hat{\rho}_b \otimes \hat{A} e^{\frac{i}{\hbar} \hat{H} t} \hat{B} e^{-\frac{i}{\hbar} \hat{H} t} \right] \\ &\approx \sum_{jk,ab} \int d\tau [\hat{\rho}_b]_w \cdot [\hat{A}]_{jk} \cdot \mathcal{T}_{ka}(t) \cdot [\hat{B}]_{ab} \cdot \tilde{\mathcal{T}}_{bj}(t),\end{aligned}\quad (53)$$

where

$$d\tau \equiv dR \cdot dP \cdot d\mathbf{x} \cdot d\mathbf{p} \cdot d\tilde{\mathbf{x}} \cdot d\tilde{\mathbf{p}} \cdot G \cdot \tilde{G},\quad (54)$$

in which $\mathbf{x} \equiv \{x_a\}$ and $\mathbf{p} \equiv \{p_a\}$ are the mapping variables associated with the forward propagator and $\tilde{\mathbf{x}}$ and $\tilde{\mathbf{p}}$ are the mapping variables associated with the backward propagator, of which both evolve according to the classical mapping Hamiltonian $\mathcal{H}(R, \mathbf{x}, \mathbf{p})$ in Eq. (52). Furthermore, $G = \exp[-\frac{1}{2} \sum_a (p_a^2 + x_a^2)]$ is the Gaussian distribution for the initial forward mapping variables, with an analogous expression \tilde{G} for backward mapping variables. In addition,

$$\mathcal{T}_{b,j}(t) = \frac{1}{2} (x_b(t) + ip_b(t))(x_j(0) - ip_j(0)),\quad (55a)$$

$$\tilde{\mathcal{T}}_{k,a}(t) = \frac{1}{2} (\tilde{x}_k(t) - i\tilde{p}_k(t))(\tilde{x}_a(0) + i\tilde{p}_a(0))\quad (55b)$$

are the transition amplitudes associated with the forward and backward propagators, respectively. The initial bath distribution is obtained by sampling from the Wigner density,

$$[\hat{\rho}_b]_w = \int d\Delta e^{-\frac{i}{\hbar} R P} \left\langle R - \frac{\Delta}{2} \left| \hat{\rho}_b \right| R + \frac{\Delta}{2} \right\rangle.\quad (56)$$

Here, R and P denote the mean positions of the bath and the momentum, respectively. The system mapping variables (for both forward and backward variables) evolve according to Hamilton's equation of motion (EOM),

$$\frac{\partial x_a}{\partial t} = \frac{\partial \mathcal{H}}{\partial p_a} = \frac{1}{\hbar} \sum_b V_{ab}(R) p_b,\quad (57a)$$

$$\frac{\partial p_a}{\partial t} = -\frac{\partial \mathcal{H}}{\partial x_a} = -\frac{1}{\hbar} \sum_b V_{ab}(R) x_b.\quad (57b)$$

The bath DOF evolves classically

$$\frac{\partial R}{\partial t} = P, \quad \frac{\partial P}{\partial t} = \mathcal{F},\quad (58a)$$

$$\mathcal{F}(R) = -\frac{1}{2} \nabla_R \{ \mathcal{H}(R, \mathbf{x}, \mathbf{p}) + \mathcal{H}(R, \tilde{\mathbf{x}}, \tilde{\mathbf{p}}) \},\quad (58b)$$

and experiences a mean force from both forward and backward mapping variables.

Note that the EOM for the mapping variables is completely equivalent to the TDSE.⁶⁸ By establishing a transformation relation,

$$x_a(t) = \sqrt{2} \cdot \Re[c_a(t)], \quad p_a(t) = \sqrt{2} \cdot \Im[c_a(t)].\quad (59)$$

Equation (58) is equivalent to the following TDSE:

$$i\hbar\dot{c}_a(t) = \sum_b V_{ab}(R(t)) \cdot c_b(t). \quad (60)$$

This was discussed by Gray and Manopolous as the symplectic integrator of TDSE.⁶⁸ For the mapping approaches with an explicit zero point energy factor (such as SQC^{62,63} and spin-LSC^{59,60}), a similar argument can also be carried out [for example, Eqs. (94) and (95) in Refs. 59 and 60]. Historically, the PLDM and related mapping dynamics approaches are often solved with a simple velocity Verlet algorithm⁷⁴ or complete symplectic algorithm^{75,76} (that requires diagonalizing \hat{V}), which are expensive to perform when having a large N in HTC Hamiltonian.

In this work, we use the relation between mapping variables and coefficients,

$$c_a = \frac{1}{\sqrt{2}}(x_a + ip_a), \quad \tilde{c}_a = \frac{1}{\sqrt{2}}(\tilde{x}_a - i\tilde{p}_a), \quad (61)$$

and the mapping EOMs in Eq. (57) become

$$i\hbar\dot{c}_a(t) = \sum_b V_{ab}(R(t)) \cdot c_b(t), \quad (62a)$$

$$-i\hbar\dot{\tilde{c}}_a(t) = \sum_b V_{ab}(R(t)) \cdot \tilde{c}_b(t). \quad (62b)$$

Here, we take advantage of the Chebyshev series in Eq. (32) and the action of \hat{H}_Q on the state vector outlined in Eq. (28), as well as the subspace decomposition and energy shift techniques outlined in Eq. (41). The nuclear force of the PLDM approach [in Eq. (58b)] for the HTC Hamiltonian is expressed as

$$\mathcal{F}(t) = -\sum_a \frac{1}{2} (|c_a(t)|^2 + |\tilde{c}_a(t)|^2) \cdot \nabla_R \langle a | \hat{H}_{sb} | a \rangle - \nabla_R \hat{H}_b, \quad (63)$$

which is computed directly using the updated coefficients $\{c_a(t), \tilde{c}_a(t)\}$ and scales linearly with \mathcal{K} . Furthermore, the transition amplitudes are also directly expressed as

$$\mathcal{T}_{bj}(t) = c_b(t) \cdot c_j^*(0), \quad (64a)$$

$$\tilde{\mathcal{T}}_{ka}(t) = \tilde{c}_k(t) \cdot \tilde{c}_a^*(0), \quad (64b)$$

which are directly computed with the forward and backward coefficients and then used to compute $C_{AB}(t)$ based on Eq. (52). For an arbitrary correlation function, $C_{AB}(t)$ still has a full sum over the indices $\{j, k, a, b\}$, which can, in principle, be further reduced if \hat{A} and \hat{B} exhibit the same type of sparsity. This is indeed the case for the linear response function (dipole–dipole autocorrelation function) with $\hat{A} = \hat{\mu}\hat{\rho}_0$ and $\hat{B} = \hat{\mu}$, as we showed in Eq. (49).

B. PLDM simulation of the linear spectra

For the linear response function expressed in Eq. (53), one can rewrite it as the sum of two correlation functions,

$$R^{(1)}(t_1) = i \text{Tr} [\hat{\mu}(0)\hat{\rho}_0\hat{\mu}(t_1)] - i \text{Tr} [\hat{\rho}_0\hat{\mu}(0)\hat{\mu}(t_1)], \quad (65)$$

where $\hat{\rho}_0 = \hat{\rho}_g \otimes \hat{\rho}_b$ [see Eq. (46)]. This can then be calculated using the expression of the PLDM correlation function [Eq. (53)]

by choosing $\hat{A} = \hat{\mu}(0)\hat{\rho}_g$, $\hat{B} = \hat{\mu}(t_1) = e^{\frac{i}{\hbar}\hat{H}t_1}\hat{\mu}e^{-\frac{i}{\hbar}\hat{H}t_1}$ and $\hat{A} = \hat{\rho}_g\hat{\mu}(0)$, $\hat{B} = \hat{\mu}(t_1) = e^{\frac{i}{\hbar}\hat{H}t_1}\hat{\mu}e^{-\frac{i}{\hbar}\hat{H}t_1}$. Alternatively, the response function can be expressed as^{25,47}

$$\begin{aligned} R^{(1)}(t_1) &= i \text{Tr} [\hat{\mu}(t_1)\hat{\mu}\hat{\rho}_g] - i \text{Tr} [\hat{\mu}(t_1)\hat{\rho}_g\hat{\mu}] \\ &= i \text{Tr} [\hat{\mu}e^{\frac{i}{\hbar}\hat{H}t_1}(\hat{\mu}\hat{\rho}_0)e^{-\frac{i}{\hbar}\hat{H}t_1}] - i \text{Tr} [\hat{\mu}e^{\frac{i}{\hbar}\hat{H}t_1}(\hat{\rho}_0\hat{\mu})e^{-\frac{i}{\hbar}\hat{H}t_1}], \end{aligned} \quad (66)$$

which can be approximated by using the PLDM approach as follows:²⁵

$$\begin{aligned} R^{(1)}(t_1) &\approx i \sum_{n_1} \sum_{n_0, \tilde{n}_0} \int d\tau_0 [\hat{\mu}\tilde{\rho}^{(1)}]_{n_1, n_1} [\hat{\mu}\hat{\rho}_g]_{n_0, \tilde{n}_0} \cdot [\hat{\rho}_b]_w \\ &\quad - i \sum_{n_1} \sum_{n_0, \tilde{n}_0} \int d\tau_0 [\hat{\mu}\tilde{\rho}^{(1)}]_{n_1, n_1} [\hat{\rho}_g\hat{\mu}]_{n_0, \tilde{n}_0} \cdot [\hat{\rho}_b]_w, \end{aligned} \quad (67)$$

where $\tilde{\rho}^{(1)}$ accounts for the transition amplitude from both forward and backward mapping trajectories, with the matrix elements expressed as follows:

$$\begin{aligned} [\tilde{\rho}^{(1)}]_{n_1, \tilde{n}_1}(t_1, n_0, \tilde{n}_0) &= \frac{1}{2} [x_{n_1}(t_1) + ip_{n_1}(t_1)] \cdot [x_{n_0}(0) - ip_{n_0}(0)] \\ &\quad \times \frac{1}{2} [\tilde{x}_{\tilde{n}_1}(t_1) - i\tilde{p}_{\tilde{n}_1}(t_1)] \cdot [\tilde{x}_{\tilde{n}_0}(0) + i\tilde{p}_{\tilde{n}_0}(0)]. \end{aligned} \quad (68)$$

The linear response at time t_1 is obtained by performing a trace over $\hat{\mu}\tilde{\rho}^{(1)}$. Using Eqs. (49) and (68), this becomes

$$\begin{aligned} &\sum_{n_1}^N [\hat{\mu}\tilde{\rho}^{(1)}]_{n_1, n_1}(t_1, n_0, \tilde{n}_0) \\ &= \sum_{n_1}^N \mu_{n_1} \cdot \frac{1}{2} [x_{n_1}(t_1) + ip_{n_1}(t_1)] \cdot [x_{n_0}(0) - ip_{n_0}(0)] \\ &\quad \times \frac{1}{2} [\tilde{x}_{\tilde{n}_1}(t_1) - i\tilde{p}_{\tilde{n}_1}(t_1)] \cdot [\tilde{x}_{\tilde{n}_0}(0) + i\tilde{p}_{\tilde{n}_0}(0)] \\ &= \sum_{n_1}^N (\mu_{n_1} \cdot c_{n_1}(t_1) \cdot c_{n_0}^*(0)) \cdot (\tilde{c}_{\tilde{n}_1}(t_1) \cdot \tilde{c}_{\tilde{n}_0}^*(0)), \end{aligned} \quad (69)$$

where in the last line we used the expression from Eq. (61).

The cavity loss dynamics is simulated using the stochastic Lindblad approach with PLDM, as detailed in Ref. 47. For a cavity mode with loss rate Γ , the Lindblad loss dynamics is equivalent to updating the forward and backward coefficients outlined in Eqs. (A3), (A5), and (A6). For any $t \in [0, t_1 - \Delta t]$, the overall propagation of the reduced density matrix for the quantum subsystem during a time step Δt can thus be summarized as

$$\hat{\rho}(t + \Delta t) = \left[e^{\mathcal{L}_L \Delta t/2} \cdot e^{\mathcal{L}_H \Delta t} \cdot e^{\mathcal{L}_L \Delta t/2} \right] \hat{\rho}(t), \quad (70)$$

where $e^{\mathcal{L}_L}$ is the decay dynamics propagation according to the stochastic Lindblad approach and $e^{\mathcal{L}_H}$ is the PLDM propagation in Eq. (62), coupled to the nuclear update with the force described in Eq. (63). In the current work, the PLDM update $e^{\mathcal{L}_H \Delta t}$ is performed using the Chebyshev series expansion outlined in Sec. III B, and the

nuclear update is performed using the velocity Verlet. The stochastic Lindblad update $e^{\mathcal{L}t}$ is performed using as outlined in Eqs. (A3), (A5), and (A6) for $\Delta t/2$ before and after the PLDM propagation. We refer to Eq. (70) as the Lindblad-Partially Linearized Density Matrix (\mathcal{L} -PLDM) approach.⁴⁷ Note that although we formally express the propagation as the reduced density matrix in Liouville space, all of the dynamical propagation has been performed in Hilbert space for the forward and backward coefficients.

To evaluate the matrix element $[\hat{\mu}\hat{\rho}_g]_{n_0, \tilde{n}_0}$, the action of the dipole operator on the ground state can be expressed as

$$\hat{\mu}\hat{\rho}_g = \sum_n \mu_n |G^0\rangle \langle E_n^0|, \quad (71)$$

which is only non-zero for N off-diagonal elements. However, the final expression of $R^{(1)}$ requires a sum of \sum_{n_0, \tilde{n}_0} . To further reduce the cost, we use the focusing algorithm and the stochastic importance sampling²⁵ to obtain the initial mapping variables, which, upon trajectory average, results in the same answer as if explicitly performing the sum \sum_{n_0, \tilde{n}_0} . For each trajectory, we use the focused initial condition for the mapping variables, with $c_{G_0} = 1$, $c_{E_i^0} = 0$ ($\forall i \in [1, N]$), $\tilde{c}_{G_0} = 0$, $\tilde{c}_{E_i^0} = 0$ ($\forall i \in [1, N]$ and $i \neq n$), and $\tilde{c}_{E_n^0} = 1$, and the focusing label n is stochastically chosen with the probability $P(n) = \mu_n / (\sum_{m=1}^N \mu_m)$ and initialize $c_{G^0} = \tilde{c}_{E_n^0} = 1$.

The first term in Eq. (67) can be computed using the following steps:

1. Sample the nuclear configuration based on $[\hat{\rho}_b]_w$. The state $|G^0\rangle \langle E_n^0|$ is randomly chosen for the focused initial condition, based on the algorithm outlined in the previous paragraph. Set the initial electronic condition $c_{G_0} = 1$ and $\tilde{c}_{E_n^0} = 1$, based on the focused initial condition, with $n_0 = G^0$ and $\tilde{n}_0 = E_n^0$, and the rest of the coefficients being zero.
2. Propagate the dynamics based on Eq. (70). In particular, for the $e^{\mathcal{L}_n \Delta t}$ part, use Eq. (62) to propagate the forward and backward electronic coefficients and Eq. (63) to compute the

nuclear force to update nuclear DOFs; for the $e^{\mathcal{L} \Delta t_k/2}$ part, use Eq. (A3) to update the coefficients to describe the Lindblad loss dynamics.

3. Compute the first term of the response function based on the estimator $i \sum_{n_1} [\hat{\mu}\hat{\rho}^{(1)}]_{n_1, n_1} \cdot [\hat{\mu}\hat{\rho}_g]_{n_0, \tilde{n}_0}$, where the initial labels of the states are $n_0 = G^0$ and $\tilde{n}_0 = E_n^0$, and explicitly compute the trace \sum_{n_1} by summing all terms expressed in Eq. (69). Average the correlation function from an ensemble of trajectories (for the $\int d\tau_0$ integral) to get the linear response signal.

For the second term, one can simply take the complex conjugate of the first term to reduce some computational costs. However, we directly compute both terms in this work (as well as in our previous work⁴⁷) and add them together. This generates $R^{(1)}(t_1)$, with examples provided in Fig. 4. Furthermore, performing the Fourier transform [see Eq. (47)] of $R^{(1)}(t_1)$ generates $R^{(1)}(\omega)$.

C. Model systems and computational details

In all simulations, we use the HTC Hamiltonian for N molecules coupled to the cavity. Each molecule is coupled to an independent phonon bath, discretized by $N_v = 20$ independent bath modes from the Debye spectral density [see Eq. (9)]. The bath parameters are sampled using the procedure outlined in Ref. 27 (or more generally, Ref. 40),

$$C_v = 2\sqrt{\frac{\lambda}{\pi N_v \omega_v} \tan^{-1}\left(\frac{\omega_{\max}}{\omega_b}\right)}, \quad (72a)$$

$$\omega_v = \omega_b \tan\left[\frac{v}{N_v} \tan^{-1}\left(\frac{\omega_{\max}}{\omega_b}\right)\right], \quad (72b)$$

where $\omega_{\max} \gg \omega_b$ is the maximum frequency when discretizing the bath frequencies. To sample within the classical regime, we choose $\omega_{\max} = 5\omega_b$. The bath parameters are $\omega_b = 18 \text{ cm}^{-1}$ (which

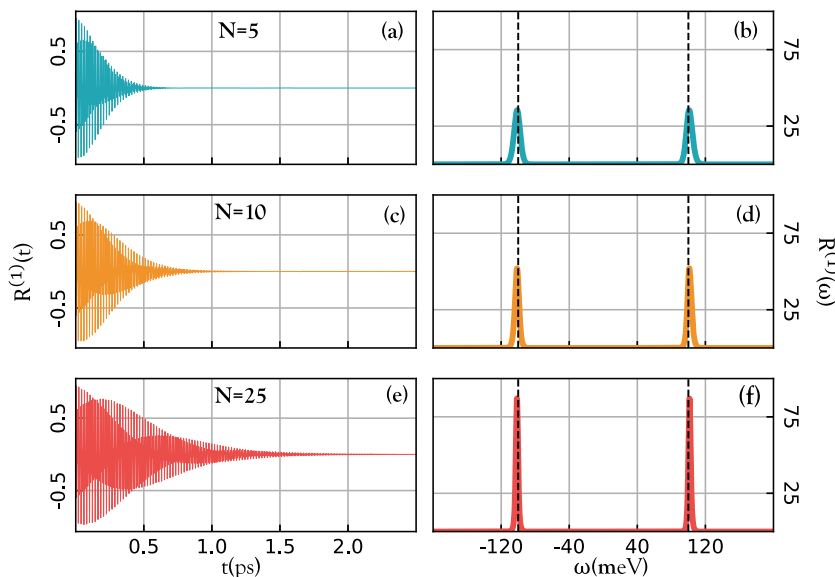


FIG. 4. Linear response function $R^{(1)}(t)$ as well as the spectra $R^{(1)}(\omega)$ of the HTC model, with different numbers of molecule N obtained from PLDM. Here, the collective Rabi splitting is fixed at $\hbar\Omega_R = 0.2 \text{ eV}$ and there is no cavity loss. Panels (a), (c), and (e) represent the time-dependent polaritonic linear response for $N = 5, 10$, and 25 molecules coupled to the cavity, respectively. Panels (b), (d), and (f) represent the normalized (by area) linear absorption spectra for $N = 5, 10$, and 25 molecules coupled to the cavity, respectively.

is 2.2 meV) and $\lambda_b = 50 \text{ cm}^{-1}$ (which is 6.2 meV). The characteristic phonon frequency ω_b leads to a bath correlation function decay timescale of 300 fs. These parameters are in the range of CdSe nanoplatelets,⁵³ which have been coupled inside the optical cavity.^{16–19} The initial bath condition $\hat{\rho}_b$ [see Eq. (56)] is described by its Wigner transform with the following analytic expression:

$$[\hat{\rho}_b]_w = \prod_{n,v} 2 \tanh\left(\frac{\beta \hbar \omega_{n,v}}{2}\right) \times \exp\left[-\tanh\left(\frac{\beta \hbar \omega_{n,v}}{2}\right) \left(\frac{\omega_{n,v}^2 R_{n,v}^2}{\hbar^2} + \frac{P_{n,v}^2}{\hbar^2 \omega_{n,v}^2}\right)\right]. \quad (73)$$

The temperature is taken to be 300 K for all simulations with $\beta = 1052.58 \text{ a.u.}$

To perform linear spectra calculations, we remain within the single excited subspace described in Eqs. (22) and (23). For a demonstration of linear response signals at different N coupled to the cavity (Fig. 4), we fix the magnitude of the collective Rabi-splitting, $\hbar\Omega_R = 2\sqrt{N}\hbar g_c = 0.2 \text{ eV}$. For the molecules, we used the average excitation energy of $\varepsilon_n = \bar{\varepsilon} = 2.0 \text{ eV}$, but to facilitate the convergence of the simulation, we shift $\bar{\varepsilon}$ to a lower value of $\bar{\varepsilon} = 0.5 \text{ eV}$. This is because of a large gap between the ground state and the excited state

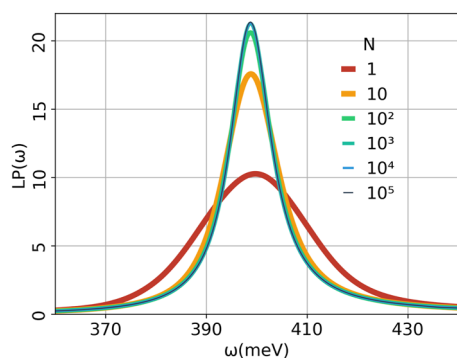


FIG. 5. Convergence of the lower polariton spectra with an increasing number of molecules from $N = 1$ to $N = 10^5$, while fixing $\hbar\Omega_R = 2\sqrt{N}$ and $\hbar g_c = 0.2 \text{ eV}$ a constant (by decreasing g_c accordingly). Here, we do not consider any type of static disorder, and the cavity loss rate is set to be $\Gamma = 20 \text{ meV}$. The phonon bath fluctuations are included.

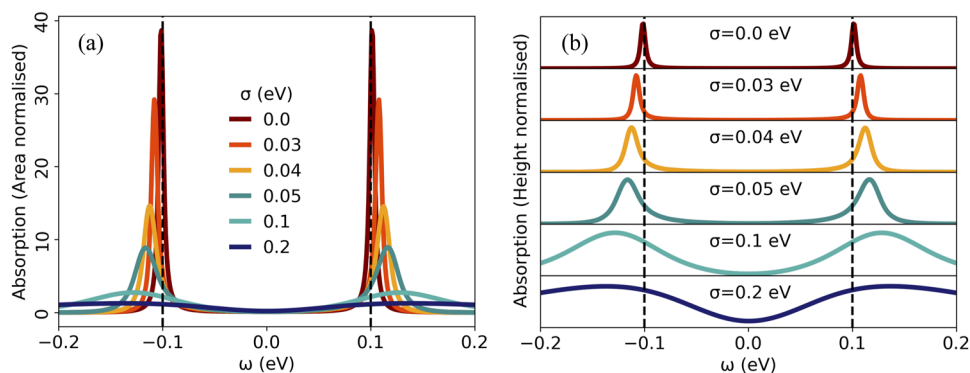


FIG. 6. Linear spectra of $N = 500$ molecules in the presence of various static energy disorders. (a) Polariton absorption spectra with each curve normalized with the corresponding area and (b) absorption curves with the peak height kept fixed. The plots are centered around the molecular excitation energy $\bar{\varepsilon} = 2.0 \text{ eV}$. The black dotted lines are the analytical lower and upper polariton peaks for a system of N molecules coupled to a single cavity mode without any energy disorder.³⁹

manifolds; the quantum transition frequency will be large, resulting in high beating frequencies in the response function. This requires a much smaller Δt in order to resolve the response function beating patterns, so the line shape (Fourier transform of response function) gives the correct absolute value of the frequencies. Meanwhile, the level-shifting trick (which is widely used in the spectra calculation literature⁷⁷) gives a much smaller energy gap from the ground state to the excited state manifold, and the quantum transitions will occur at a lower frequency, allowing us to use a much larger Δt . In Appendix C, we discuss the details of this approximation. The cavity frequency is also kept at $\hbar\omega_c = 0.5 \text{ eV}$ (zero light–matter detuning). The cavity is lossless with $\Gamma = 0 \text{ meV}$. This trick is applied in Figs. 4 and 5 and not in Figs. 6 and 7.

To simulate disorder effects in linear spectra (Figs. 6 and 7), we fix the collective coupling strength $2\sqrt{N}\hbar g_c$ at 0.2 eV for $N = 500$ molecules. For the molecules, we used the average excitation energy of $\bar{\varepsilon} = 2.0 \text{ eV}$. Throughout this work, we consider the resonance condition where

$$\bar{\varepsilon} = \hbar\omega_c, \quad (74)$$

where we have ignored the solvent reorganization energy λ_b [see Eq. (9)] in the above condition due to its small value. For Fig. 6, ε_n is sampled based on a Gaussian distribution [Eq. (75)], while $\bar{\varepsilon} = 2.0 \text{ eV}$. For results presented in Fig. 7, there is no site energy disorder and $\varepsilon_n = \bar{\varepsilon} = 2.0 \text{ eV}$. The cavity frequency is thus kept at $\hbar\omega_c = 2.0 \text{ eV}$ (zero light–matter detuning). The cavity loss rate was set at $\Gamma = 10 \text{ meV}$. The nuclear time step for Verlet propagation was taken to be $\Delta t = 10 \text{ a.u.}$ ($\approx 0.25 \text{ fs}$), and the forward and backward coefficients are evolved between nuclear propagation using Eq. (32).

For all results presented here, we always consider dynamical disorders due to the exciton–phonon couplings. For results presented in Figs. 6 and 7, we further consider the influence of static disorders. For energy static disorder simulations (Fig. 6), the excitation energy for the n th molecule is sampled from the Gaussian distribution,

$$P(\varepsilon_n) = \frac{1}{\sqrt{2\pi\sigma_\varepsilon^2}} e^{-(\varepsilon_n - \bar{\varepsilon})^2 / 2\sigma_\varepsilon^2}, \quad (75)$$

where σ_ε is the width of the Gaussian centered at $\bar{\varepsilon}$. This introduces a static inhomogeneity within the molecular excitation energies. The sampling is performed independently for each trajectory.

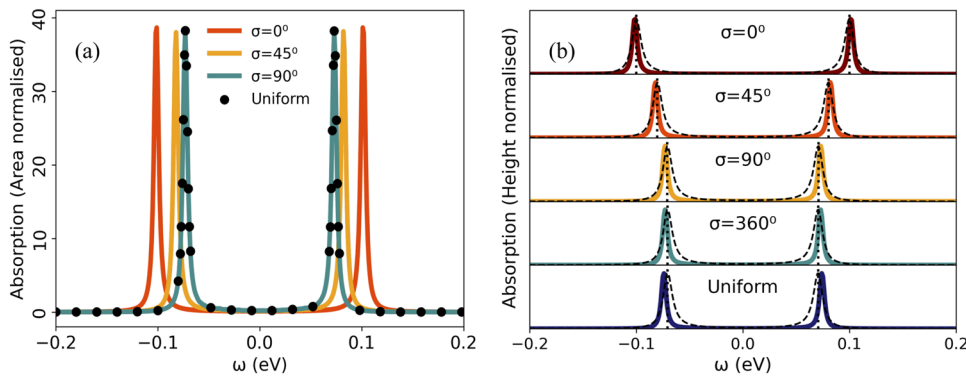


FIG. 7. Linear spectra of $N = 500$ molecules coupled to the cavity, in the presence of dipole orientation disorders. (a) Polariton absorption spectra with each curve normalized by area and (b) the same absorption spectra normalized by peak height. The plots are centered around the molecular excitation energy $\varepsilon_0 = \bar{\varepsilon} = 2.0$ eV. The black dashed lines are the analytic result using Eq. (81). The vertical dotted lines indicate the effective Rabi splitting estimated from Eq. (82).

Following the previous work,³⁸ we also study the effect of orientation disorder by sampling the molecular orientation (with respect to the cavity mode) from the Gaussian distribution,

$$P(\theta_n) = \frac{1}{\sqrt{2\pi\sigma_\theta^2}} e^{-(\theta_n - \bar{\theta})^2 / 2\sigma_\theta^2}. \quad (76)$$

The disorders in $\{\theta_n\}$ will influence the molecule-cavity coupling in Eq. (6). In the limit of $\sigma_\theta \rightarrow \infty$, it corresponds to an isotropic distribution of $\theta_0 \in [0, \pi]$ relative to the polarization direction of the cavity field. The expectation value of $\cos^2 \theta$ for the Gaussian distribution is

$$\langle \cos^2 \theta \rangle = \int_{-\infty}^{\infty} d\theta P(\theta) \cdot \cos^2 \theta = \frac{1}{2} + \frac{1}{2} e^{-2\sigma_\theta^2}. \quad (77)$$

Following the previous work,³⁸ we further assume that the external laser field $\mathbf{E}(t)$ in the linear spectra simulation has the same direction as the cavity polarization direction $\hat{\mathbf{e}}$ such that

$$\hat{\boldsymbol{\mu}} \cdot \mathbf{E}(t) = \sum_n \mu_n \cos \theta_n \cdot E(t). \quad (78)$$

This means that the $\hat{\boldsymbol{\mu}}$ expression in $R^{(1)}(t_1)$ when considering dipole orientational disorders will be expressed as

$$\hat{\boldsymbol{\mu}} = \sum_n \mu_n \cos \theta_n. \quad (79)$$

In this work, we assume that the dipole will remain in a two-dimensional plane and deviate from the field polarization direction $\hat{\mathbf{e}}$.

To ensure the tight convergence of $R^{(1)}(t)$, we used a total of 10^6 trajectories, although typically a total of $10^3 - 5 \times 10^3$ trajectories will provide visually converged results. The propagation time step for the Chebyshev propagation was chosen as $\Delta t = 20$ a.u. (≈ 0.5 fs). Once having a converged $R^{(1)}(t)$, we perform the numerical Fourier transform,

$$R^{(1)}(\omega) = \sum_{j=0}^{T/\Delta t} \Delta t \cdot e^{i\omega j \Delta t} R^{(1)}(j \Delta t) \cos\left(\frac{\pi \cdot j \Delta t}{2T}\right),$$

where T is chosen such that $R^{(1)}(t)$ is sufficiently decay to zero. For Fig. 4, for all N , the response signal was simulated up to $T = 2.5$ ps,

where $R^{(1)}(t) \rightarrow 0$. In Fig. 5, all response signals were simulated till $T = 800$ fs. In Fig. 6, the response signal for each energy disorder [Eq. (75)] decays to 0 with different T . For no disorder case, $\sigma_\varepsilon = 0.0$ eV, we chose $T = 1.5$ ps. For $\sigma_\varepsilon = 0.03, 0.04, 0.05, 0.1, 0.2$ eV, we chose $T = 1$ ps, $T = 500$ fs, $T = 300$ fs, $T = 80$ fs, and $T = 50$ fs, respectively. To simulate the dipole disorder [Eq. (76)], we set $T = 1.5$ ps for all σ_θ cases in Fig. 7.

V. RESULTS AND DISCUSSIONS

A. Linear response signal

Figure 4 presents the time-dependent linear response signal, $R^{(1)}(t)$, and the linear absorption spectra, $R^{(1)}(\omega)$, simulated using the PLDM approach. In this figure, we do not account for any type of disorder, and we keep $\sqrt{N}\hbar g_c$ a constant as we change N . Figures 4(a) and 4(b) present the linear response signal and the linear absorption spectra for $N = 5$ molecules coupled to the cavity, respectively. For the $R^{(1)}(\omega)$ data, we shift the axis with respect to $\hbar\omega_0 = \bar{\varepsilon}$, so the spectra are centered around zero instead of $\bar{\varepsilon}$. A clear observation of the linear response signal indicates the presence of two types of oscillations that arise due to the two polaritonic transitions, $|G^0\rangle \rightarrow |-\rangle$ and $|G^0\rangle \rightarrow |+\rangle$, as observed in the two peaks in linear spectra at $\hbar\omega - \bar{\varepsilon} \approx -100$ meV and $\hbar\omega - \bar{\varepsilon} \approx 100$ meV, respectively. Due to symmetry, there is no optical transition from $|G^0\rangle$ to the dark state manifold $\{|D_k\rangle\}$.

Figures 4(c) and 4(d) correspond to the case for $N = 10$ molecules. Since the linear response signal is a direct measurement of the coherent oscillations between the upper and lower polariton states, compared to panel (a), we can observe an increase in the coherence lifetime in $R^{(1)}(t)$ for the case of $N = 10$, which manifests itself as a further linewidth reduction and an increase in peak height for linear spectra $R^{(1)}(\omega)$, as shown in panel (d). This is due to the well-known polaron decoupling effect,^{1,2,78} which effectively reduces the phonon reorganization energy by $\mathcal{O}(1/N)$ under the collective coupling regime [see Eq. (27) in Ref. 50 and the discussions there]. Figures 4(e) and 4(f) present the results of coupling $N = 25$ molecules to the cavity mode, and we observe a further enhancement in the response signal lifetime and a further reduction in the polariton linewidths.

In Fig. 5, we present the convergence of absorption line shape with an increasing number of molecules from $N = 1$ to $N = 10^5$,

while fixing the collective Rabi splitting, $\hbar\Omega_R = 2\sqrt{N}\hbar g_c = 0.2$ eV. We use a cavity loss rate of $\Gamma = 20$ meV and the same molecular parameters as used in Fig. 4. Figure 5 presents the lower polariton line shape for a different number of molecules. The LP spectra with $N = 10^2$ molecules (green curve) are visually close to the converged line shape of $N = 10^5$ molecules (thin dark blue curve). From these curves, the line shape features of linear spectroscopy due to collective effects appear to reach a convergence at $N \approx \mathcal{O}(10^2)$, which is very much closer to the line shape features of $N = 10^5$. The theoretical approaches of this work allow us to directly perform the quantum dynamics and linear spectra simulations with $N = 10^5$ molecules and explore various trends of convergences.

B. Energy disorder

Figure 6 presents the linear absorption spectra of $N = 500$ molecules collectively coupled to the cavity, where each molecule is experiencing static energy disorder (inhomogeneous disorder), described in Eq. (75). Figure 6(a) presents the absorption normalized with respect to the area, and Fig. 6(b) presents the absorption normalized with respect to the intensity of the peak, with the vertical black dashed lines indicating the Rabi splitting without any static site energy disorders.

As the disorder in energy σ_ϵ approaches the same size of Rabi-splitting $\hbar\Omega_R = 2\sqrt{N}\hbar g_c$, one can observe that the effective Rabi-splitting slightly increases. This has been understood from perturbation theory,^{38,79} where the effective Rabi splitting is expressed as $\hbar\Omega_R \approx 2\sqrt{N}\hbar g_c + \langle \delta\epsilon \rangle^2 / (\sqrt{N}\hbar g_c)$ when $\sigma_\epsilon / (\sqrt{N}\hbar g_c) \ll 1$ (see Appendix C of Ref. 38), as well as the approximate analytical solution of Rabi splitting when having Gaussian-type energy disorders.⁷⁹

As the linewidth increases with an increase in inhomogeneous disorder, the peak intensity decreases. This is because the effective number of molecules with zero detuning decreases. When further increasing σ_ϵ to 0.2 eV [dark blue curve in Fig. 6(a)], the UP and LP peaks start to merge and the Rabi-splitting fades away. This pattern is clearer when we normalize the curves with the highest peak intensity, as shown in Fig. 6(b). These observations also agree with the recent theoretical studies.^{38,79,80}

C. Dipole orientation disorder

Figure 7 presents the linear spectra of $N = 500$ molecules coupled to a single cavity mode under various dipole orientation disorders based on Eq. (76), without any static energy disorders, $\epsilon_n = \bar{\epsilon} = \epsilon$. We further center the spectra by setting $\hbar\omega_0 = \epsilon = 0$. As one gradually increases the dipole orientation disorder σ_θ , Rabi splitting decreases. Meanwhile, the linewidth of the polariton peaks remains almost unchanged, since there are no additional broadening mechanisms when including dipole disorders, as opposed to the previous case when considering an inhomogeneous energetic disorder in the system.

For the resonance condition, $\hbar\omega_c = \bar{\epsilon}$, when considering the dipole orientation disorder, the polariton eigenspectrum is still analytically available.³⁷ By setting $\omega_0 = \omega_c$, one can use the well-known analytic results of the line shape⁸¹ to the expression of absorption spectra as follows:

$$R^{(1)}(\omega) = \frac{\Gamma \kappa N g_c^2 \langle \cos^2 \theta \rangle}{\left| (\omega - \omega_c + i \frac{\Gamma}{2\hbar}) (\omega - \omega_0 + i \frac{\kappa}{2\hbar}) - N g_c^2 \langle \cos^2 \theta \rangle \right|^2}, \quad (80)$$

where κ is the molecular linewidth⁴⁷ and $\langle \cos^2 \theta \rangle$ [see Eq. (77)] is the effective modification of polariton frequencies depending on orientation disorder.^{37,38}

In Fig. 7(b), we present the numerical results obtained from our simulations (solid colors) when normalized to the peak height, as well as the analytic answer from Eq. (80) (black dashed lines). Overall, the analytic answer gives a reasonable estimate of the line shape behavior across different angle disorders but slightly overestimates the linewidth. The narrower linewidth from the numerical results is likely due to the subaverage behavior^{50,82,83} [linewidth smaller than $\Gamma/2 + \kappa/2$ for the resonant case, see Eq. (34) in Ref. 50], which cannot be captured by the simple expression of the line shape in Eq. (80) that predicts the linewidth to be $\Gamma/2 + \kappa/2$ under the resonance case.

In Fig. 7(b), we see that with an increase in dipole orientation disorder σ_θ , the effective Rabi-splitting decreases until it converges to a specific Ω_R . The convergence is reached around a disorder of $\sigma_\theta = 90^\circ$. This can be interpreted as follows. For a HTC model Hamiltonian under the condition $\hbar\omega_c = \bar{\epsilon} = \epsilon$, the polariton energies can be expressed as^{37,39}

$$\epsilon_\pm = \epsilon \pm \frac{1}{2} \sqrt{4N\hbar^2 g_c^2 \langle \cos^2 \theta \rangle + \left(-i \left[\frac{1}{2} \kappa - \frac{1}{2} \Gamma \right] \right)^2}, \quad (81)$$

where $\langle \cos^2 \theta \rangle$ is expressed in Eq. (77) and λ_b is the molecular reorganization energy [cf. Eq. (9)], which accounts for the additional light-matter detuning when the cavity frequency is equal to the bare molecular excitation. For the model system considered here, which is in the slow bath limit (inhomogeneous limit),⁸⁴

$\kappa/2 = \sqrt{\ln 2} \cdot \sqrt{\frac{\lambda_b k_B T}{\hbar}}$, and $\sqrt{\ln 2}$ is the half of full-width half maximum (FWHM) conversion factor for a Gaussian line shape. The real part of Eq. (82) gives the location of the polariton peaks, and the imaginary part gives the linewidth of the respective polaritons.^{39,47}

For the resonance case, the Rabi splitting is expressed as

$$\hbar\Omega_R = \text{Re}[\epsilon_+ - \epsilon_-], \quad (82)$$

where ϵ_\pm can be evaluated from Eq. (81). One can estimate the Rabi splitting and compare with the numerically simulated spectra. Indeed, the Rabi splitting decreases as one increases the dipole disorder from the fully aligned case (when $\sigma_\theta \rightarrow 0$ and thus $\langle \cos^2 \theta \rangle = 1$) to the fully isotropic case in the 2D plane (when $\sigma_\theta \rightarrow \infty$ and thus $\langle \cos^2 \theta \rangle = 1/2$). The vertical dashed line in each panel of Fig. 7(b) indicates the Rabi splitting based on the corresponding value of $\langle \cos^2 \theta \rangle$ [cf. Eq. (77)], and the numerical simulations agree well with the analytic result. The same behavior has also been discussed in Appendix D of Ref. 38.

VI. CONCLUSION

In this paper, we presented two general theoretical techniques for an efficient simulation of polariton quantum dynamics under the collective light-matter coupling regime. The first one is related to the sparsity of the HTC Hamiltonian, which allows one to reduce the cost of acting polariton Hamiltonian onto a state vector to the

linear order of the number of states, instead of the quadratic order. In principle, a direct density-matrix-based simulation requires a computational cost (in both time propagation and memory requirements) proportional to $\mathcal{O}(N^6)$, where N is the number of molecules and $\mathcal{O}(N^2)$ is the number of states in the double excitation subspace. The sparse nature of the HTC system Hamiltonian and dipole matrix allow for a compact expression of $\hat{H}_Q|\Psi_s\rangle$ and $\hat{\mu}|\Psi_s\rangle$ as simple Hadamard products between vectors of $\mathcal{O}(N^2)$ instead of matrix–matrix or matrix–vector operations.

The theoretical technique applies the well-known Chebyshev series expansion approach for quantum dynamics propagation and to simulate the polariton dynamics in the HTC system. The Chebyshev expansion approach allows us to use a much larger time step for propagation and only requires a few recursive operations of the polariton Hamiltonian acting on state vectors, which we can further take advantage of the first technique we developed in this work. Due to the block diagonal structure of the HTC-type Hamiltonian, one can decompose Chebyshev expansions for single and double excitation subspaces separately, and a shift in the energy axis of the subspace significantly reduces the oscillation magnitude of the coefficients in time and allows a much larger propagation time step. These two theoretical approaches are general and can be applied to any trajectory-based non-adiabatic quantum dynamics methods.

Applying these two theoretical techniques, we extended the \mathcal{L} -PLDM method⁸⁵ to the collective coupling regime. Using this approach, we simulate the linear spectra of the N -molecule polaritonic system described by the HTC Hamiltonian. The approach we present here will significantly reduce the computational costs. To demonstrate the developed methods, we simulate the polaritonic absorption spectra of $N = 500$ molecules coupled to a cavity under either static energy disorder or dipole orientation disorder, under which the permutation symmetry of the HTC Hamiltonian is broken and a simple mean-field approach is not directly applicable. We observe similar trends of these linear spectra as observed in the previous work.^{79,80,86} As the disorder in energy σ_e increases, the linewidth increases, and the Rabi splitting also gradually increases due to the disorders in energies. When gradually increasing the orientation disorder, the Rabi splitting decreases, without a further change in the linewidth. In future work, we will outline the theoretical details of stimulating 2DES spectra for the polariton system under the collective coupling regime (Paper II), with new computational techniques for focusing the algorithm of the PLDM simulation. Going beyond the single cavity mode approximation, the theoretical development presented here can be applied to investigate many molecules collectively coupled to many cavity modes inside a Fabry–Pérot cavity, such as the transport quantum dynamics⁸⁷ or angle-resolved photoluminescence spectra.^{16,17}

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AUTHOR DECLARATIONS

Conflict of Interest

The authors have no conflicts to disclose.

Author Contributions

M. Elious Mondal: Conceptualization (equal); Data curation (lead); Formal analysis (lead); Investigation (lead); Methodology (lead); Software (lead); Validation (lead); Visualization (lead); Writing – original draft (lead); Writing – review & editing (equal). **A. Nickolas Vamivakas:** Conceptualization (supporting); Funding acquisition (equal); Investigation (supporting); Project administration (equal); Supervision (supporting); Writing – review & editing (supporting). **Steven T. Cundiff:** Conceptualization (equal); Formal analysis (equal); Funding acquisition (equal); Investigation (supporting); Project administration (equal); Supervision (supporting); Writing – review & editing (supporting). **Todd D. Krauss:** Conceptualization (equal); Formal analysis (equal); Funding acquisition (lead); Investigation (equal); Methodology (equal); Project administration (lead); Supervision (equal); Validation (supporting); Writing – review & editing (supporting). **Pengfei Huo:** Conceptualization (equal); Formal analysis (equal); Funding acquisition (equal); Investigation (equal); Methodology (equal); Project administration (lead); Supervision (lead); Writing – original draft (equal); Writing – review & editing (lead).

DATA AVAILABILITY

The data that support the findings of this work are available from the corresponding author on reasonable request.

APPENDIX A: CAVITY LOSS THROUGH LINDBLAD

The loss channel from state $|G^1\rangle$ to state $|G^0\rangle$ can be described by the loss operator

$$\hat{L} = |G^0\rangle\langle G^1|. \quad (\text{A1})$$

The dissipator $\mathcal{L}_{\hat{L}}$ accounts for the cavity loss channel causing the system to relax,

$$\mathcal{L}_{\hat{L}}[\hat{\rho}_Q] = \Gamma \left(\hat{L} \hat{\rho}_Q \hat{L}^\dagger - \frac{1}{2} \{ \hat{L}^\dagger \hat{L}, \hat{\rho}_Q \} \right), \quad (\text{A2})$$

where Γ is the rate of relaxation of the jump operator, which quantifies the coupling strength of the system to the environment, $\{\hat{A}, \hat{B}\} = \hat{A}\hat{B} + \hat{B}\hat{A}$ is the anti-commutator, and $\hat{\rho}_Q = \text{Tr}_b[\hat{\rho}]$ is the reduced density matrix operator for the quantum subsystem by tracing out all bath DOF. In this study, Γ is the cavity loss rate, and the cavity quality factor is defined as $Q = \hbar\omega_c/\Gamma$.

The Lindblad evolution for the forward and backward wavefunctions is computed by multiplying the forward and backward coefficients with a phase factor for these two states as

$$c_a(t + dt) = \eta_a(t)c_a(t), \quad (\text{A3})$$

$$\tilde{c}_a(t + dt) = \tilde{\eta}_a(t)\tilde{c}_a(t), \quad (\text{A4})$$

where the phase factors are

$$\eta_{G^1} = \tilde{\eta}_{G^1} = e^{-\Gamma dt/2}, \quad (\text{A5})$$

$$\eta_{G^0} = \tilde{\eta}_{G^0}^* = 1 + (2r - 1) \sqrt{3(1 - e^{-\Gamma dt}) \frac{c_{G^1}(t) \tilde{c}_{G^1}^*(t)}{c_{G^0}(t) \tilde{c}_{G^0}^*(t)}}, \quad (\text{A6})$$

and $r \in [0, 1]$ is a uniform random number. The phase factor, when averaged over the ensemble of trajectories, ensures the exponential population decay of the state $|G^1\rangle\langle G^1|$ and a gain in the population of the state $|G^0\rangle\langle G^0|$ as required by Lindblad dynamics, as analytically proved in our previous work.^{47,88} The rest of the forward and backward coefficients, for $a \neq G^1$ and $a \neq G^0$, are not affected by the Lindblad decay, and the corresponding $\eta_a = \tilde{\eta}_a = 1$. When averaging among a reasonable amount of trajectories (typically $\sim 10^2$ trajectories, see Appendix B in Ref. 47), Eqs. (A5) and (A6) exactly reproduce the Lindblad EOM for the reduced density matrix, as demonstrated numerically in our previous work.⁴⁷

APPENDIX B: ERROR ANALYSIS FOR VARIOUS PROPAGATORS

We estimate the error accumulated during a propagation step Δt in the quantum subspace for various propagation schemes in Fig. 8. We first compute the eigenstates of the HTC Hamiltonian and compute the exact propagator by

$$\exp\left(-\frac{i\hat{H}_Q\Delta t}{\hbar}\right) = \hat{U}^\dagger \exp\left(-\frac{i\Delta t}{\hbar} \hat{\Lambda}\right) \hat{U}, \quad (\text{B1})$$

where $\hat{\Lambda}$ is the eigenvalue matrix (the diagonal form of \hat{H}_Q) and \hat{U} is the eigenvector matrix, satisfying $\hat{H}_Q\hat{U} = \hat{\Lambda}\hat{U}$. The evolution of any vector can now be obtained by operating Eq. (B1) on the vector. To test the accuracy of the different propagators, we generated $\mathcal{N} = 10^6$ samples of random state vectors by randomly selecting the coefficients in Eq. (26) (in the complex plane) while keeping it properly normalized. For each configuration, the bath fluctuation term $\sum_v C_{n,v} R_{n,v}$ in \hat{H}_Q [see Eq. (27)] is also randomly sampled, based on

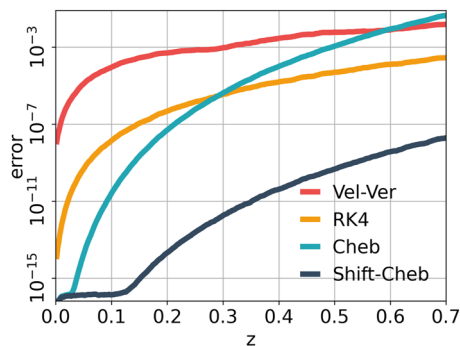


FIG. 8. Accuracy of different types of propagators as a function of z -value [see Eq. (33)]. The red, yellow, cyan, and dark blue curves correspond to the symplectic integrator (velocity Verlet), RK4, Chebyshev, and shifted Chebyshev propagators. The z -value scales linearly with Δt according to $z = 13.6 \cdot \Delta t$, and Δt is taken in fs. So, $z = 0.1$ corresponds to $\Delta t = 1.36$ fs.

the Wigner distribution [Eq. (73)]. For each value of the uncertainty parameter z , we compute the error of the propagator by comparing it against the exact evolution by

$$\text{error} = \frac{1}{\mathcal{N}} \sum_{v=1}^{\mathcal{N}} \max \{|c_i^v - c_i^{\text{ex}}|\} \quad \forall i \in [1, \mathcal{K}], \quad (\text{B2})$$

where c_i^{ex} are the components of the exactly propagated vector through Eq. (B1) for the v th trial and c_i^v are the components of the vector propagated by the other numerical integrators.

APPENDIX C: SHIFTING THE CENTRAL FREQUENCY OF DIFFERENT EXCITATION MANIFOLDS

Note that the HTC Hamiltonian in Eq. (5) assumes the rotating wave approximation and ignores the counter-rotating wave terms $\sum_n \hbar g_c^n (\hat{\sigma}_n^\dagger \hat{a}^\dagger + \hat{\sigma}_n \hat{a})$ and dipole self-energy terms. See Refs. 39 and 89 for details. For the physical model we considered here, $\hbar\omega_c = 2.0$ eV, and the largest $\sqrt{N}g_c = 100$ meV such that $\sqrt{N}g_c/\hbar\omega_c = 0.05 < 0.1$, and the system does not enter into the ultra-strong coupling regime (defined as $0.1 < \sqrt{N}g_c/\hbar\omega_c < 1.0$), and we expect that the counter-rotating wave terms and dipole self-energy terms do not significantly impact⁸⁹ the polariton eigenenergy and spectra.

On top of this, we have introduced a computational trick in Sec. IV C, which provides a constant shift for a given manifold of the energy. This allows a much larger Δt in the simulation and thus reduces the computational costs. We emphasize that the choice to

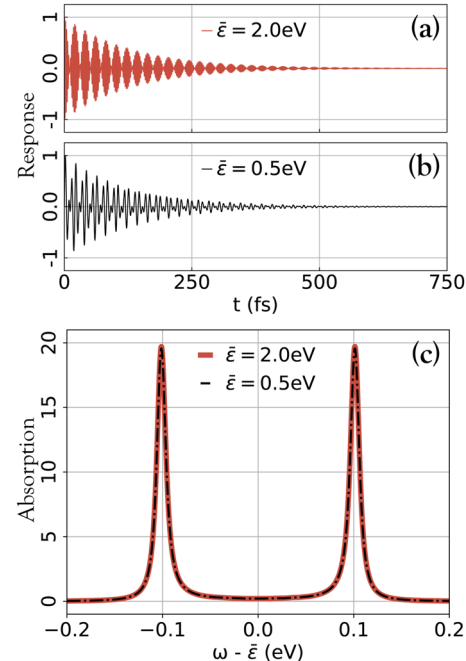


FIG. 9. Effect of shifting the central excitation frequency of the system, on the linear spectra of polaritons. Panels (a) and (b) describe the time-dependent linear response for the original ($\bar{\epsilon} = 2.0$ eV) and shifted ($\bar{\epsilon} = 0.5$ eV), respectively. Panel (c) compares the resulting absorption spectra from both systems.

shift down the exciton energy to 0.5 eV is for the purpose of reducing computational cost and will generate the same spectra (in terms of the relative peak positions and shape of the peaks) as for our original system with $\omega_0 = 2.0$ eV, of course, with a shifted absolute value of peak frequency.

Figure 9 shows the results to demonstrate the shifting idea, where we use $N = 100$ molecules, a cavity loss rate of $\Gamma = 20$ meV, and effective Rabi splitting of $\Omega_R = 200$ meV. There are no static disorders considered in these results. We have kept all the other parameters the same, except shifting $\bar{\epsilon} = 2.0$ eV [Fig. 9(a)] to $\bar{\epsilon} = 0.5$ eV [Fig. 9(b)]. As one can see, there is more frequent beating in the response function in Fig. 9(a), due to the 2 eV optical frequency. In Fig. 9(b), the response function beats with a much smaller frequency, due to the reduced optical gap of only 0.5 eV. As such, a smaller Δt is needed for Fig. 9(a) due to the high-frequency beating.

Figure 9(c) presents the line shape obtained from the Fourier transform of the response functions. One can see that the line shapes are identical for both cases, including the splitting magnitude and the width and shape for each peak. The only difference is that for $\bar{\epsilon} = 2.0$ eV, the spectra are centered at 2 eV, and for $\bar{\epsilon} = 0.5$ eV, the spectra are centered at 0.5 eV.

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