Hybrid cluster-expansion and density-functional-theory approach for optical absorption in TiO$_2$

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A combined approach of first-principles density-functional calculations and the systematic cluster-expansion scheme is presented. The dipole, quadrupole, and Coulomb matrix elements obtained from ab initio calculations are used as an input to the microscopic many-body theory of the excitonic optical response. To demonstrate the hybrid approach for a nontrivial semiconductor system, the near-bandgap excitonic optical absorption of rutile TiO$_2$ is computed. Comparison with experiments yields strong evidence that the observed near-bandgap features are due to a dipole-forbidden but quadrupole-allowed 1s-exciton state.

I. INTRODUCTION

The single-particle electronic states are the basis needed to quantitatively model the excitonic optical properties of a semiconductor system. In many cases, we can use semiempirical methods like the k·p theory together with the effective-mass approximation to obtain the electronic wave functions and energies. However, there are many systems which are not characterized well enough for the needed input parameters to be reliably known. These kinds of "nontrivial" systems are found, for example, in novel semiconductor materials containing more than two constituents, like ternary, quaternary and even more complex compounds, in dilute bismides or nitrides, in organic systems, in organic/inorganic heterostructures, and in complex interfaces.

Even seemingly simple binary systems such as bulk rutile TiO$_2$ pose considerable challenges. At first sight, TiO$_2$ seems to be a common semiconductor material that has been widely used in applications and intensively studied over several decades. Despite this, after a second look, one quickly realizes that TiO$_2$ clearly belongs into the class of nontrivial systems mentioned above, because many of its essential parameters are poorly known. For example, the reported conduction-band effective mass in rutile varies almost two orders of magnitude. Additionally, it exhibits an exceptionally large refractive index in the visible range as well as particularly strong birefringence and dispersion properties.

Furthermore, rutile has highly asymmetric dielectric properties regarding different crystallographic directions with a particularly large magnitude of the low-frequency dielectric constant. This results in an exceptionally strong screening of the Coulomb interaction between electrons and holes for excitons with a binding energy below or comparable to the polar phonon energies. Nevertheless, optical absorption measurements yields strong evidence that the observed near-bandgap features are due to a dipole-forbidden but quadrupole-allowed 1s-exciton state.

near the band gap of TiO$_2$ have shown some indications of excitonic features which, however, are very controversially discussed in the literature. For example, the excitonic signatures were interpreted as a dipole-forbidden but quadrupole-allowed 1s exciton, or as a weakly dipole-allowed 2p exciton state.

In order to study the optical properties of rutile and to develop a general scheme that allows us to overcome the restrictions of the semiempirical models, we have to use more systematic methods to access the microscopic properties of the electronic states. Here, the most widely used scheme is density functional theory (DFT) which has been proven to be a very efficient approach in obtaining ground-state properties for various solid-state systems, molecules, nanostructures, liquids, and molecules adsorbed on surfaces.

The DFT computed single-particle energies and wave functions can then be used as an input to the cluster-expansion scheme that provides first-principles level description of many-body dynamics. Its strength relies on a systematic grouping of different correlations as clusters that build up sequentially in time. The cluster-expansion method is widely applied in semiconductor solid-state systems and it can even describe quantum-optical properties as well as strongly interacting Bose gas quantitatively.

In this work, we combine a DFT approach with the cluster-expansion method to obtain a hybrid cluster-expansion and density-functional-theory (CE-DFT) scheme for studying dynamical many-body quantum phenomena. We first obtain the needed Coulomb and optical matrix elements via DFT for the second-quantization system Hamiltonian that we then use to model dynamical effects by applying the cluster-expansion scheme. The electronic excitation dynamics are derived in a very general form, yielding a structure similar to the semiconductor Bloch equations. Prior to our work, ab initio matrix elements have been used together with the semiconductor Bloch equations...
to compute the surface exciton properties on a Si surface [39]. Linear optical properties such as the absorption follow directly once the dynamics have been solved. We then apply our combined CE-DFT method for rutile TiO$_2$ and study its near-bandgap optical properties by considering electric-dipole, electric-quadrupole, and magnetic-dipole light–matter interactions. We also show that the electric-quadrupole interaction is highly dependent on the propagation and polarization directions of the light. We use our microscopic results to analyze the light–matter and electron–electron interactions, respectively.

II. THEORETICAL BACKGROUND

To effectively formulate the quantum kinetics of a semiconductor system via the cluster-expansion method, we start from a generic Hamiltonian [3, 28]

$$H = \sum_i E_i a_i^\dagger a_i - E(t) \sum_{i,j} F_{ij} a_i^\dagger a_j$$

$$+ \frac{1}{2} \sum_{i,j,i',j'} V_{i'j'i,j} a_i^\dagger a_i a_j^\dagger a_j,$$

(1)

where $a_i^\dagger$ ($a_i$) is the creation (annihilation) operator of an electronic state defined by the particle index $i$ and $E(t)$ is the time-dependent electric field. This Hamiltonian is uniquely defined by $E_i$, $F_{ij}$, and $V_{i'j'i,j}$, i.e., by the energies of the electronic states and the matrix elements for light–matter and electron–electron interactions, respectively.

To identify these input parameters from a DFT calculation, we repeat the derivation of Eq. (1) from the Lagrangian and Hamiltonian formalism of classical electrodynamics [40]. In the Coulomb gauge, the quantization yields the conventional minimal-substitution Hamiltonian for electronic quasiparticles [3, 11]:

$$H_N = \sum_{i=1}^N \left\{ \frac{1}{2m_0} \left[ p_i - eA(r_i,t) \right]^2 + U(r_i) \right\}$$

$$+ \frac{1}{2} \sum_{i\neq j} V(r_i,r_j),$$

(2)

where $m_0$ is the free-electron mass, $e$ is the charge of an electron, $A(r,t)$ is the vector potential of an optical field, $U(r)$ is an external potential, and $V(r,r')$ is a pairwise electron–electron interaction.

Equation (2) includes a contribution proportional to the square of the optical field that can be eliminated via the Power-Zienau-Woolley transformation [40, 42] in our initial Coulomb-gauge Lagrangian. This way, we obtain a classical Hamiltonian where the light–matter interaction is directly given by the electric $E(r,t)$ and the magnetic $B(r,t)$ fields instead of the vector potential. After quantization, it reads

$$H' = \sum_{i=1}^N H(p_i, r_i) - E(t) \sum_{j=1}^N F(p_i, r_i)$$

$$+ \frac{1}{2} \sum_{i\neq j} V(r_i, r_j) + H_{RM},$$

(3)

where we assumed fields of the form $E(t, r) = E(t)E(r)$ and $B(t, r) = E(t)B(r)/\omega$ with the optical frequency $\omega$. The diamagnetic contributions [40, 42] $H_{RM}$ can be neglected at moderate field intensities, hence we omit this term in the following. The single-particle Hamiltonian is given by

$$H(p, r) = \frac{p^2}{2m_0} + U(r),$$

(4)

and the light–matter-interaction operator by

$$F(p, r) = -e\mathbf{r} \cdot \sum_{n=0}^{\infty} \frac{1}{(n+1)!} \left[ (r \cdot \nabla \phi')^n E(r') \right]_{r'=0}$$

$$- \frac{e}{2m_0} \sum_{l=0}^{\infty} \frac{l+1}{(l+2)!} \left\{ \mathbf{p} \cdot \left[ r \times (r \cdot \nabla \phi')^l B(r') \right] \right\}$$

$$+ \left[ r \times (r \cdot \nabla \phi')^l B(r') \right] \cdot \mathbf{p},$$

(5)

after expressing $E(r)$ and $B(r)$ through their multipole expansions. The electric-dipole interaction ($n = 0$) can be weak, especially near the direct band gap of TiO$_2$ [14, 21]. To keep our approach general while simultaneously suitable for near-bandgap optical properties of TiO$_2$, we initially consider all terms within the multipole expansions of $E(r)$ and $B(r)$.

If we know the electronic wave functions $\phi_i(r)$ and energies $E_i$ that follow from the Schrödinger equation

$$H(p, r) \phi_i(r) = E_i \phi_i(r),$$

(6)

we can construct the fermionic field operator

$$\Psi(r) = \sum_i \phi_i(r) a_i,$$

(7)

As elaborated in many text books [2, 43], we can use $\Psi(r)$ in the second-quantization step for $H(p, r), F(p, r)$, and $V(r, r')$ to obtain the Hamiltonian [11] from Eq. (6) with the matrix elements

$$F_{ij} = \int d^3 r \phi_i^*(r) F(p, r) \phi_j^*(r),$$

(8)

$$V_{i'j'i,j} = \int d^3 r \int d^3 r' \phi_i^*(r) \phi_j^*(r') V(r, r') \phi_{i'}(r') \phi_{j'}(r).$$

(9)

In principle, $\phi_i(r)$ could be chosen to be any complete orthonormal set for the second-quantization step. In practice, it is useful if they are eigenfunctions of a suitable effective Hamiltonian that is known to give eigenvalues $E_i$ corresponding to accurate ionization energies and
electron affinities. A possible choice is a DFT Hamiltonian with a scissor shift to obtain the correct band gap, even better would be to use a scissor shifted static GW Hamiltonian if it is available (we avoid explicit frequency dependence in the single-particle Hamiltonian in order to keep the time evolution simple and ensure orthogonal wave functions). The unshifted Hamiltonian can be written by replacing the potential \( U(\rho_0) \) with

\[
U[\rho_0](\vec{r}) = V_{\text{ext}}(\vec{r}) + V_{\text{H}}[\rho_0](\vec{r}) + V_{\text{xc}}[\rho_0](\vec{r}),
\]

where the external potential from the nuclei \( V_{\text{ext}}(\vec{r}) \) is augmented with an effective potential that includes the interaction of the electrons with the average electron density \( V_{\text{H}}[\rho_0](\vec{r}) \), as well as approximate exchange and correlation effects \( V_{\text{xc}}[\rho_0](\vec{r}) \). Note that these terms are explicitly dependent on the equilibrium density matrix \( \rho_0 \) that is determined in DFT by a self-consistent iterative procedure. To correct the band gap we shift the eigenvalues by a downshift for the states below the Fermi level and an upshift on the states above it. The terms we have added to \( U(\vec{r}) \) include some Coulomb terms that must be later subtracted when we develop the equations of motion as we discuss in Appendix A.

One of the most critical conditions for the success of our CE-DFT method is that the computed \( \phi_\nu(\vec{r}) \) and \( E_\nu \) are sufficiently close to the actual one-particle wave function and energies, respectively. When we assume that this condition is fulfilled, by utilizing Eqs. (5) and (10), we obtain a DFT-defined system Hamiltonian \( H_i \) that is the basis of our hybrid CE-DFT approach.

### III. CE-DFT EQUATION FOR OPTICAL ABSORPTION

It is beneficial to divide the electronic index \( i \) in Eq. (11) into two parts \( i \rightarrow (\lambda, j) \) where \( \lambda = v \) or \( \lambda = c \) for states that are either occupied or unoccupied in the ground state of the system, respectively, and \( j \) defines the exact state inside these groups. After we have constructed our DFT-defined system Hamiltonian \( H_i \), we can study the dynamics of the expectation values \( P_{ij}^{\lambda\nu} \equiv \langle a_{\lambda;i}^\dagger a_{\lambda;j} \rangle \) via the Heisenberg equation of motion and the cluster-expansion method \( \mathcal{E} \) \( \mathcal{E} \). This yields the semiconductor Bloch equations

\[
\frac{i\hbar}{\partial t} P_{ij}^{\lambda\nu} = \sum_k \left[ \hat{E}_{kj}^{\lambda} P_{ik}^{\lambda\nu} - \hat{E}_{ik}^{\lambda} P_{kj}^{\lambda\nu} + \Omega_{ik}^{\lambda} \hat{P}_{kj}^{\lambda\nu} - \Omega_{kj}^{\lambda} \hat{P}_{ik}^{\lambda\nu} \right] + \Gamma_{ij}^{\lambda\nu},
\]

where \( \hat{E}_{ij}^{\lambda} (\Omega_{ij}^{\lambda}) \) is the renormalized kinetic energy (Rabi frequency) and \( \tilde{\lambda} \) denotes the complement of \( \lambda \) (\( \tilde{c} = v \)). Explicit forms of \( \hat{E}_{ij}^{\lambda} \) and \( \Omega_{ij}^{\lambda} \) can be found in Appendix A. The term \( \Gamma_{ij}^{\lambda\nu} \) includes the coupling to the higher-order clusters that follows from the two-particle contributions \( \mathcal{E} \) \( \mathcal{E} \) due to Coulomb and phonon interactions.

Similarly to the formulation of Eq. (11), we can construct equations for higher order clusters and study, e.g., dynamics of exciton correlations \( \mathcal{E} \) \( \mathcal{E} \). However, in this work we focus on fundamental aspects of combining DFT and the cluster-expansion method, thus limiting our considerations on the level of Eq. (11).

In this situation, Eq. (11) yields a closed set of equations between the dynamics of the microscopic polarization \( P_{ij} \equiv P_{ij}^{\nu} \), the electron occupation \( f_{ij}^\nu = \hat{P}_{ij}^{\nu} \), and the hole occupation \( f_{ij}^\nu = \delta_{ij} - P_{ij}^{\nu} \). Even though this set of equations can be used to model many interesting dynamical effects such as nonlinearities or excitation induced changes in the optical response, in the present study we concentrate on the linear absorption properties of the rutile TiO\(_2\) system.

In this case, \( f_{ij}^\nu = f_{ij}^{\nu*} = 0 \) in Eq. (11) and we can neglect several terms that are connected to optical nonlinearities. When we also make the Tamm-Dancoff approximation by omitting processes that would generate coupling between the \( P_{ij}^{\nu} \) and \( F_{ij}^{\nu} \) terms, Eq. (11) reduces to the (see Appendix A for details)

\[
\frac{i\hbar}{\partial t} P_{ij} = \left[ E_j^\nu - E_i^\nu \right] P_{ij} - \sum_{k,l} V_{kj;li}^{\nu;\nu;\nu} P_{kl} - E(t) F_{ji}^{\nu} + \Gamma_{ij}^{\nu},
\]

containing

\[
V_{\lambda_1,\lambda_2;\lambda_3,\lambda_4}^{\nu_1,\nu_2,\nu_3,\nu_4} \equiv V_{\lambda_1,\lambda_2;\lambda_3,\lambda_4}^{\nu_1,\nu_2,\nu_3,\nu_4} - V_{\lambda_2,\lambda_1;\lambda_3,\lambda_4}^{\nu_1,\nu_2,\nu_3,\nu_4},
\]

where we have grouped the direct and exchange-electron–electron interactions under a single matrix element. Here, we use a notation where the \( \lambda \) index in the matrix elements is written as a superscript above the corresponding \( i \) index, e.g., \( E_i^\nu \equiv E_{\nu,i} \). While Eq. (11) can be used for any system defined by the Hamiltonian \( H \) and when rotating-wave contributions dominate the dynamics, Eq. (12) is applicable only in cases where the Tamm-Dancoff approximation is justified.

The homogeneous solution of Eq. (12) defines the Wannier equation \( \mathcal{E} \) \( \mathcal{E} \)

\[
[E_j^\nu - E_i^\nu]\phi_\nu(i,j) - \sum_{k,l} V_{kj;li}^{\nu;\nu;\nu} \phi_\nu(k,l) = E_\nu \phi_\nu(i,j),
\]

as an eigenvalue problem for the wave function \( \phi_\nu(i,j) \) and the energy \( E_\nu \) of an exciton state \( \nu \). Converting the polarization into the exciton basis,

\[
P_{ij} = \sum_\nu p_\nu \phi_\nu(i,j), \quad p_\nu = \sum_{i,j} \phi_\nu^*(i,j) P_{ij},
\]

the Fourier transform of Eq. (12) yields

\[
[h\omega + i\gamma_\nu(\omega)]p_\nu(\omega) = E_\nu p_\nu(\omega) - F_\nu E(\omega),
\]

where \( E(\omega) \) is the Fourier transform of \( E(t) \) and

\[
F_\nu \equiv \sum_{i,j} \phi_\nu^*(i,j) F_{ji}^{\nu}
\]
defines a generalized oscillator strength of optical transitions.

To avoid a detailed analysis of Coulomb and phonon scattering, we introduce a dephasing function

$$\gamma_\lambda(\nu) = \frac{\gamma_\nu}{\exp[(E_\nu - \hbar \omega - \mu_\nu)/\Delta E] + 1}$$  \hspace{1cm} (18)

that phenomenologically includes the contributions of the $\Gamma_\nu^\nu$ term in Eq. (10). With the parameters $\gamma_\nu$, $\mu_\nu$, and $\Delta E_\nu$, function $\gamma_\nu(\omega)$ describes the excitation-induced dephasing effects [28, 44] as well as the exponential decay of the absorption tail towards lower energies; a phenomenon known as the Urbach tail [2, 42], which is of particularly significance in polar semiconductors [46] like TiO$_2$.

We can now solve Eq. (10) directly, yielding

$$p_\nu(\omega) = \frac{e_\nu}{E_\nu - \hbar \omega - i\gamma_\nu(\omega)} E(\omega).$$  \hspace{1cm} (19)

The absorption response of our system then follows from the imaginary part of the linear susceptibility $\chi(\omega) = \frac{p_{\nu}(\omega)}{P(\omega)}$ where $P(\omega) = \sum_{\nu} F_\nu^\nu p_\nu(\omega)$ is the macroscopic polarization. With the help of Eq. (19), the Elliott formula becomes

$$\alpha(\omega) = \frac{\omega}{n c} \text{Im}[\chi(\omega)] = \frac{\omega}{\varepsilon_0 n c} \sum_{\nu} \frac{|F_\nu|^2 \gamma_\nu(\omega)}{(E_\nu - \hbar \omega)^2 + \gamma_\nu^2(\omega)},$$  \hspace{1cm} (20)

where $n$ is the refractive index of the material.

IV. EXCITONIC ABSORPTION IN RUTILE

A. Two-band model

To apply the general results of the previous section for TiO$_2$, we assume a three-dimensional crystal having an infinite volume. Then, the potential $U(r)$ in the Hamiltonian (1) must have the periodicity of the lattice, and we can use the Bloch theorem to produce electronic wave functions of the form

$$\phi_{\lambda,k}(r) = \frac{e^{i k \cdot r}}{(2\pi)^{3/2}} u_{\lambda,k}(r),$$  \hspace{1cm} (21)

where $u_{\lambda,k}(r)$ is a lattice-periodic Bloch function, $k$ is the wave vector, and $\lambda$ denotes different Bloch bands. Thus, in the main step before constructing the system Hamiltonian (1) of our crystal, we solve the band structure $E_k^\lambda$ and the Bloch functions $u_{\lambda,k}(r)$ corresponding to the Schrödinger equation (5) via DFT, requiring the form of Eq. (21).

Our DFT results presented in Sec. [IV] reveal that it is well justified to model the near-bandgap optical properties of TiO$_2$ by including only the energetically lowest conduction band $\lambda = c$ and the highest valence band $\lambda = v$. Hence, we can relate the Bloch-band index $\lambda$ and the $\lambda$ index of Sec. [III] and directly use Eqs. (11)- (17) if we replace the sums over the particle index by $k$-space integrals. Furthermore, in the vicinity of the band gap $E_g$ we find (see Sec. [IV]) that the electron energy $E_k^c \equiv E_k^c - E_g$ and the hole energy $E_k^h \equiv -E_k^h$ are accurately described by a parabolic dispersion

$$E_k^\lambda = \frac{\hbar^2}{2m_\lambda^c} k_x^2 + \frac{\hbar^2}{2m_\lambda^v} k_y^2 + \frac{\hbar^2}{2m_\lambda^v} k_z^2,$$  \hspace{1cm} (22)

where $m_\lambda^c$ is the $i$-directional effective mass of an electron or hole for $\lambda = e$ or $\lambda = h$, respectively.

B. Light–matter interaction

When we express $E(r)$ and $B(r)$ in Eq. (5) via their multipole expansions, the matrix element in Eq. (8) is related to the position operator $r$. In our crystal system, the position operator $r$ maps a wave function by $r \phi_{\lambda,k}(r)$ out of the Hilbert space that is constructed by the lattice-periodic wave functions (17). Consequently, the $r$ matrix element is not uniquely defined and, for example, in DFT computations its value depends on how we chose the unit cell of a system (18). This is a known problem that has been widely studied in connection with the electric-dipole matrix element over several decades (48, 51) and is still an important research topic (52).

In general, while the light–matter matrix element in Eq. (8) is not directly observable, it manifests itself through other quantities like the oscillator strength $F_\nu$. By considering $F_\nu$ in Eq. (17) for an infinite crystal under the assumption that a two-band model with a parabolic band structure is valid, we present in Appendix [B] a method where the ambiguities related to the $r$ matrix elements can be avoided by expressing the light–matter matrix elements via the momentum matrix element $p_k^\lambda \lambda'$. Using our DFT Bloch functions, these matrix elements are computed via

$$p_k^\lambda \lambda' = \frac{1}{\Omega_0} \int_{\Omega_0} d^3 r u_{\lambda,k}^*(r) p u_{\lambda',k}(r),$$  \hspace{1cm} (23)

with the unit cell volume $\Omega_0$.

Even though we discuss in Appendix [B] how our approach can be used to solve $F_\nu$ for arbitrary $E(r, t)$ and $B(r, t)$, we focus on the electric-dipole, electric-quadrupole, and magnetic-dipole light–matter interactions. Furthermore, we assume that a plane wave with $E(r) = e^{i q \cdot r} \mathbf{E}_g$ and $B(r) = e^{i q \cdot r} (q \times \mathbf{E}_g)$ propagates through our sample, where $q \mathbf{E}_g$ and $\mathbf{E}_g$ define the wavevector and the polarization direction of the optical field, respectively. Then, the $n = 0$, $n = 1$, and $l = 0$ terms of
Eq. (5) yield the light–matter interaction operators
\[ D \equiv - e \mathbf{e}_E \cdot \mathbf{r}, \]
\[ Q \equiv - i \frac{e}{\hbar} \left( \mathbf{e}_E \cdot \mathbf{r} \right) (\mathbf{q}_E \cdot \mathbf{r}), \]
\[ M \equiv - \frac{e}{2\omega n_0} [\mathbf{r} \times (\mathbf{q}_E \times \mathbf{e}_E)] \cdot \mathbf{p}, \]
for the electric-dipole, electric-quadrupole and magnetic-dipole interactions, respectively.

By using the steps outlined in Appendix C we connect the operators \( D, Q, \) and \( M \) to the matrix elements
\[ D_{kk'} = \delta(k - k') D_{kk'}, \]
\[ Q_{kk'} = \delta(k - k') \sum_{i,j} e_i Q_{k;ij}^c q_j, \]
\[ M_{kk'} = \delta(k - k') \sum_{i,j} e_i M_{k;ij}^c q_j. \]

Here, \( e_i \) and \( q_i \) are the \( i \)-directional components of the vectors \( \mathbf{e}_E \) and \( \mathbf{q}_E \), respectively. \( D_{kk'}^c = - ie \mathbf{e}_E \cdot \mathbf{p}_k^c \) includes the conventional \([2, 3, 18]\) relation between the \( \mathbf{r} \) and \( \mathbf{p} \) matrix elements via
\[ \mathbf{p}_k^\lambda' \equiv \frac{\hbar}{m_0} \mathbf{E}_k^\lambda' - E_k^\lambda, \]
and the explicit forms of \( Q_{k;ij}^c \) and \( M_{k;ij}^c \) are given in Appendix C. The matrix element \( F_{kk'}^{cv} \) becomes
\[ F_{kk'}^{cv} = \delta(k - k')(D_{kk'}^c + Q_{k;ij}^c + M_{k;ij}^c). \]

### C. Coulomb interaction and direct excitonic states

For rutile TiO\(_2\), it is necessary to model the electron–electron interaction \( V(\mathbf{r}, \mathbf{r}') \) using the result for anisotropic media \([5, 13]\). The resulting expression can be given through the Fourier transformation
\[ V(\mathbf{r}, \mathbf{r}') = \int d^3k V_k e^{i \mathbf{k} \cdot (\mathbf{r} - \mathbf{r}')}, \]
\[ V_k = \frac{e^2}{8\pi^3 \varepsilon_0} \frac{1}{\varepsilon_k k_0^2 + \varepsilon_k k_2^2 + \varepsilon_k k_3^2}, \]
where \( \varepsilon_0 \) is the vacuum permittivity, \( \varepsilon_k \) is the dielectric constant along the \( i \)-directional principal axis of the permittivity tensor \([5, 13]\), and \( k_i \) is the related component of the \( \mathbf{k} \) vector, respectively. If our system has an excitation in a sufficiently small region of the Brillouin zone, we can quite generally approximate the Coulomb matrix element by
\[ V_{k_1,k_2,k_3,k_4}^{\lambda_1,\lambda_2,\lambda_3,\lambda_4} = \delta(k_1 + k_2 - k_3 - k_4) \delta_{\lambda_1,\lambda_4} \delta_{\lambda_2,\lambda_3} V_{k_1 - k_4}, \]
which follows from Eqs. (9), (32), and (33) (see Appendix C and Ref. [21]). By using \( u_{c,k}(r) \) and \( v_{c,k}(r) \) in TiO\(_2\) determined by DFT computations, we carefully check in Appendix C that the relevant Coulomb interaction is indeed well approximated by the matrix element \([54]\).

Since the light–matter and Coulomb interactions conserve the crystal momentum \( \hbar \mathbf{k} \) via the matrix elements in Eqs. (31) and (34), we find from Eq. (11) that only direct quantities \( F_{kk'}^{\lambda\lambda'} = F_{kk'}^{\lambda'\lambda} \) couple to optical excitation if we omit possible nondiagonalities in \( \Gamma_{kk'}^{\lambda} \). Consequently, we need to consider only the direct excitonic states with wave functions \( \phi_{\nu}(\mathbf{k}) = \phi_{\nu}(\mathbf{k}, \mathbf{k}) \) for the optical response following form Eq. (15). These states are solutions of the Wannier equation
\[ (E_k - E_{k_0}) \phi_{\nu}(\mathbf{k}) = \int d^3k' V_{k-k'} \phi_{\nu}(\mathbf{k'}) = E_{\nu} \phi_{\nu}(\mathbf{k}), \]
where \( \mathbf{k} \) can be associated with the relative motion of electrons and holes, and we have assumed that the approximations in Eqs. (22) and (23) are valid \([3, 20]\). In Eq. (35), the \( \nu \) index typically refers to the symmetry of the exciton states with values \( \nu = 1s, 2p, \) etc. In an anisotropic system, we can assume that this symmetry based grouping is only approximate and the character of, e.g., s and p states becomes mixed.

Equation (35) can have simultaneously anisotropic energies \( E_k^\lambda \) and matrix elements \( V_k^c \), which may complicate solution strategies. However, we can always perform a \( k \)-space coordinate transform in Eq. (35) that results in a transformed spherical Coulomb matrix element (see Appendix D). The use of \( V_{k_{1234}}^{\lambda_1,\lambda_2,\lambda_3,\lambda_4} \) in Eq. (31) together with this coordinate transformation comprises a rather general and remarkably beneficial approach for modeling properties of anisotropic systems. This is due to the fact that DFT can be effectively used to compute \( E_k^\lambda \) and \( F_{kk'}^{\lambda\lambda'} \), whereas computing all the Coulomb-interaction terms would be a numerically highly demanding task; especially in three-dimensional problems. Furthermore, a spherical Coulomb matrix element makes it possible to solve the excitonic states and the related properties of a system efficiently by using techniques developed elsewhere \([5, 14]\). In particular, we expand the excitonic wave functions \( \phi_{\nu}(\mathbf{k}) \) and all matrix elements in terms of spherical harmonics in the transformed coordinate system and then solve the corresponding high-dimensional eigenvalue problem given by Eq. (35) as discussed in Appendix D.

### D. Band structure

In our numerical evaluations, we first set-up our computational cell based on the experimentally determined \([18]\) tetragonal crystal structure of rutile TiO\(_2\) that is defined by the unit cell with lattice constants \( a = b = 4.59 \) Å and \( c = 2.96 \) Å. The inset of Fig. 1(a) depicts the unit cell in a Cartesian coordinate system where oxygen (titanium) atoms are illustrated by the red (gray) spheres.

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Starting from this unit cell, we perform a DFT calculation with the SIESTA [55] code using the PBE exchange-correlation functional [56], Trouiller-Martins pseudopotentials [57], and a DZP basis set of atomic-like orbitals with an energy shift parameter of 100 meV. After a sufficient convergence of the ground-state properties (with 11x11x17 k points), we rediagonalize the Kohn-Sham Hamiltonian matrix at the k points we are interested in, obtaining the needed $E^\lambda_k$ and $u_{\lambda,k}(r)$.

To compute the matrix elements in Eqs. (8) and (9), we put the cell-periodic part of the wave functions on a real space grid and Fourier transform them to obtain their plane-wave representation. All needed matrix elements can then be computed either in real or reciprocal space. We use the DFT computed energies to obtain the effective masses, while the light–matter-interaction matrix elements in Eq. (8) are used directly.

Our computed Coulomb matrix elements in Eq. (10) were found to correspond well to the matrix element (34) for the small ($k_1 - k_2$)-values that are important for excitons and we thus chose to use the latter for simplicity. In addition of confirming the approximation (34), the ab initio Coulomb matrix elements were used in a phase matching procedure to obtain the light–matter-interaction matrix elements that correspond to the phase fixing we do when we adopt the matrix element (34). Following this procedure, we obtain all needed matrix elements for Eq. (10), which establishes the technical basis of our CE-DFT approach in TiO$_2$.

The actual $E^\lambda_k$ dispersion obtained from our DFT computations is shown in Fig. 1(a) along the high-symmetry points of the Brillouin zone. The depicted band structure agrees to some extent with multiple previous results [58–60] and indicates a direct gap at the Γ point. Our DFT computations is shown in Fig. 1(a) along the high-symmetry points of the Brillouin zone. The depicted band structure and its parabolic approximation. (a) A wide region of the ab initio band structure throughout the whole Brillouin zone, where the k-space region of interest is indicated by the shaded areas. (b) Electron (solid lines) and hole (dashed lines) energies for $k_\lambda = 0$ in the region of interest as a function of $k_x$ (black lines) and $k_z$ (red lines). Parabolic fits for electron and hole energies (lines) approximate ab initio energies (squares and circles) to a high degree of accuracy.

The shaded rectangles in Fig. 1(a) indicate the electronic states most important for near-bandgap optical transitions, covering roughly a range of 20 meV in electron–hole energy, $E^e_k - E^h_k$, around the Γ point. In this region, the energetically lowest conduction and highest valence band are relatively well separated from the nearby bands such that we can adopt the two-band approach described in Sec. IV A. Figure 1(b) shows the electron (continuous lines and squares) and the hole energies (dashed lines and circles) near the Γ point as a function of $k_z$ (black lines) and $k_z$ (red lines) when $k_y = 0$. The reciprocal $k_x$, $k_y$, and $k_z$ directions correspond to lattice x, y, and z directions, respectively. These directions are selected to be parallel to the a, b, and c axes of TiO$_2$ crystal structure, as indicated in the insert of Fig. 1(a).

The computed ab initio $E^e_k$ and $E^h_k$ energies are almost parabolic in all directions and symmetric with respect to $k_x \leftrightarrow k_y$ exchange. However, $k_x$ (or $k_y$) and $k_z$ directions exhibit a different $k$ dependence, which creates a strong anisotropy. This anisotropy can be accurately described by Eq. (22) with effective masses $m^e_x = 0.13 m_0$, $m^e_y = 0.519 m_0$, $m^e_z = 3.8 m_0$, and $m^h_z = 5.3 m_0$. Figure 1(b) compares the parabolic model (lines) and the ab initio energies (squares and circles), showing the high accuracy of our approximation. In this figure, the wave vector is scaled by the exciton Bohr radius $a_0 = 4\pi \epsilon_0 e^2/\mu$ with geometrically averaged reduced mass $\mu \equiv (\mu^e x^e z^h)^{1/3}$ and dielectric constant $\epsilon \equiv (\epsilon^e z^h)^{1/3}$, where $\epsilon_x = \epsilon_y = 111$ and $\epsilon_z = 257$. 
E. Dipole and quadrupole matrix elements

Figure 2(a) shows the imaginary part of the dipole matrix element $D_k^{cv}$ that we obtain from Eq. (27) with the DFT defined Bloch functions. The real part of $D_k^{cv}$ is negligible. We have selected a $x$-polarized light mode, $\mathbf{e}_E \parallel \hat{x}$, and then computed $D_k^{cv}$ along the $k_x$ axis (black line), $k_y$ axis (red line), and direction $k(\hat{k}_x + \hat{k}_y)$ (blue line). In the $k$-space region of interest, the dipole matrix element becomes almost linearly dependent on $k_x$. The black line indicates that this dependency is not completely linear due to the visible nonlinearity for $|k a_0| > 3$. In the $k_{xy}$ direction (blue line), $D_k^{cv}$ remains linear even far away from the Γ point.

It follows from the linear-in-$k_x$-like dependency that $D_k^{cv}$ remains small along the $k_y$ axis (red line) and includes only a minor dependency on $k_x$ near the Γ point. The dipole matrix element for the $y$-polarized light mode, $\mathbf{e}_E \parallel \hat{y}$, is essentially equal to the $\mathbf{e}_E \parallel \hat{x}$ matrix element with a sign change and with respect to the $k_x \leftrightarrow k_y$ transformation. In the immediate vicinity of the Γ point, our results indicate that the dipole matrix element for $z$-polarized light modes, $\mathbf{e}_E \parallel \hat{z}$, is negligible, yielding a vanishing near-bandgap absorption in TiO$_2$ for this particular light polarization, as already found in several works [13, 21].

In Fig. 2(b), we show the absolute value of the $Q_{k,xx}^{cv}$ element of the quadrupole-matrix tensor multiplied by the wave number $q = E_g n_0 / (hc)$ where $c$ is the speed of light in vacuum and $n_0 = 2.95$ is the ordinary refractive index of TiO$_2$, corresponding to a light mode polarized in the $xy$ plane. We have used the same color scheme as in Fig. 2(a). In the region of interest, the $g|Q_{k,xx}^{cv}|$ remains constant along the $k_{xy}$ direction while it has a parabolic behavior along the $k_x$ and $k_y$ axes.

In the vicinity of the Γ point, all other components of the quadrupole and magnetic-dipole matrix-element tensors are at least two orders of magnitude smaller than $Q_{k,xx}^{cv}$. The only exception is the $Q_{k,yy}^{cv}$ counterpart of $Q_{k,xx}^{cv}$ (obtained from $Q_{k,xx}^{cv}$ via the $k_x \leftrightarrow k_y$ transformation). Due to their small magnitude, the other elements play only a minor role for the near-bandgap optical properties of TiO$_2$. Nevertheless, we still include them in our numerical computations. The large difference in the magnitudes of the different components of the $Q_{k,xx}^{cv}$ tensor indicates that the near-bandgap quadrupole light–matter interaction in TiO$_2$ should be highly dependent on the propagation and polarization directions of light. Due to the complicated nature of the electric-dipole, electric-quadrupole, and magnetic-dipole matrix elements, we take them as a direct input from the DFT without any approximations.

F. Optical absorption

Figure 3(a) shows the optical absorption spectrum following from Eq. (29) when we use the low-frequency values $\epsilon_\perp = 111$ and $\epsilon_{||} = 257$ to describe the dielectric screening of Coulomb interaction between electrons and holes in Eq. (33). We have chosen the polarization of light to be always in the $xy$ plane whereas we change the polar $\theta$ and azimuthal $\varphi$ angles that are the angles of propagation with respect to the $z$ and $x$ axes of the unit cell, respectively. Here, we present the absorption spectrum for four sets of $(\theta, \varphi)$ combinations.

Whenever light propagates in the $x, y,$ or $z$ planes, e.g., in the cases denoted by $(\theta = 0, \varphi = 0)$ (shaded area) and $(0, \pi/4)$ (black line) in Fig. 3(a), the $Q_{k,xx}^{cv}$ and $Q_{k,yy}^{cv}$ elements of the quadrupole matrix-element tensor do not contribute to the light–matter interaction. Consequently, the absorption spectrum then practically follows only from the electric-dipole interaction. Since the $D_k^{cv}$ matrix elements for $x$- and $y$-polarized light are almost linear in $k_x$ and $k_y$, respectively, the electric-dipole in-
teraction couples light predominantly to the $p_{xy}$-like excitonic states of the $xy$ plane \[14, 15, 21\]. From the $xy$-plane symmetry of our system, it follows that these $p_{xy}$ states are degenerate and by changing the angles $\theta$ and $\varphi$ we cannot drastically change their combined contribution to the optical response, which is shown by the comparison of the spectra labeled $(0, 0)$ and $(0, \pi/4)$ in Fig. 3(a). Thus, the major propagation- and polarization-angle dependent changes in our absorption spectrum (for $xy$ polarization) must follow from the quadrupole interaction.

In Fig. 3(a), we also show two spectra denoted as $(\pi/2, \pi/8)$ (blue line) and $(\pi/2, \pi/4)$ (red line) in which the propagation direction points away from the $x$, $y$, or $z$ planes. When comparing the spectra $(0, 0)$ and $(\pi/2, \pi/8)$, we see that while the overall absorption intensity is slightly increased, a weak resonance at the spectral position of the $1s$ exciton appears. This transition is dipole-forbidden but quadrupole-allowed \[14, 17, 21\]. These trends are further strengthened when the azimuth angle is increased to $\varphi = \pi/4$, yielding the $(\pi/2, \pi/4)$ set with the strongest quadrupole resonance.

In Fig. 3(b), we study how the peak intensity of the $1s$ state changes as a function of the angles $\theta$ and $\varphi$ that give the spectra labeled as $(\theta, \pi/4)$ (black line) and $(\pi/2, \varphi)$ (red line), respectively. More generally, whenever $\theta$ ($\varphi$) is varied we find that $\varphi = \pi/4$ ($\theta = \pi/2$) gives the strongest $1s$ intensity. As a summary, we show the angle dependency of the $1s$-peak intensity in Fig. 3(b), varying only one angle at a time while fixing the other. The shape of these curves follows almost identically the corresponding angular dependency of $|Q_{00}|^2$.

Even with the most optimal angle combination $(\pi/2, \pi/4)$ for the quadrupole interaction, the resulting excitonic absorption signature is still weaker than the experimentally observed feature \[14, 15, 21\]. This is explained by the extremely small binding energy of $E_{1s}^B = 0.5$ meV ($E_{2p_{xy}} = 0.1$ meV) for the $1s$ ($2p_{xy}$) exciton due to the unusually large dielectric constants $\epsilon_\perp = 111$ and $\epsilon_\parallel = 257$. Additionally, the experimentally measured continuum absorption intensity around 2 meV above the band gap has been reported to be between 25 cm$^{-1}$ and 50 cm$^{-1}$ \[14, 18, 21\], i.e., our result differ from this roughly by a factor of five.

Generally, it is not obvious whether we should use a low-frequency, a high-frequency or some other phenomenological value of a dielectric constant when we model optical excitations in semiconductors \[20\]. This is of particular significance in TiO$_2$ where the difference between high-frequency constants \[18\] ($\epsilon_\infty = 6.8$ and $\epsilon_\parallel^\infty = 8.4$) and low-frequency constants ($\epsilon_\perp^0 = 111$ and $\epsilon_\parallel^0 = 257$) is exceptionally large. In rutile, the contributions of the different sources to the screening of the Coulomb interaction, e.g., phonons and the ionic nature of TiO$_2$ \[21\], are poorly known \[15\].

If we adjust the components of the dielectric constant between the high and low frequency values via $\epsilon_i(w) = \epsilon_i^0 - (\epsilon_i^0 - \epsilon_i^\infty)w$, where $w$ is between 0 and 1, we can change the binding energies of excitons so that they become visible in the optical spectra. In the inset of Fig. 3(b), we show how the binding energies of $1s$ (black line) and $2p$ (red line) change as a function of $\epsilon(w) = [\epsilon_i^0(w)\epsilon_i^\infty(w)]^{1/3}$. By selecting either $\epsilon(w) = 24.3$ or $\epsilon(w) = 47.8$, we can tune the absorption to show a distinct resonance 4 meV below the band gap. Effectively, we have adjusted the binding energies of either the $2p_{xy}$ or $1s$ state being the origin of that signature.

In Fig. 3 we compare absorption spectra for different $(\theta, \varphi)$ sets of propagation angles and dielectric constants to the available experiments. If we choose $(0, 0)$ and $\epsilon(w) = 24.3$ (black line in Fig. 3(a)), the experimental resonance from Ref. \[15\] (shaded area) is assigned to the $2p$ exciton. However, the general shape of the spec-
in the form of Eq. (22), $\epsilon(w)$ and the effective masses become connected via Eq. (35). If we change the value of $\epsilon(w)$, we not only describe the screening of electron–electron interaction phenomenologically but also correct possible errors in our effective masses. Second, without considering $\epsilon(w)$, we can assume that our Coulomb matrix element is highly accurate and cannot contribute to this problem, as discussed in Appendix C. Hence, the only remaining uncertainty is in the light–matter matrix elements. Thirdly, since both the 2p-resonance and the continuum absorption (dominantly) originate from the same source (dipole interaction), possible errors in the dipole matrix element $D_\kappa^w$ would lead only to the scaling of both features with the same factor without a change of the general shape of the spectrum. Thus, based on our results it seems to be practically impossible to model the experimentally observed excitonic feature via 2p states.

No such problems occur if we model the excitonic resonance as the 1s state by selecting $\epsilon(w) = 47.8$, yielding $E_{1s}^B = 4.0$ meV and $E_{2p_x}^B = 1.0$ meV. Interestingly, the experimentally obtained [14, 21] binding-energy difference between 1s and 2p states of 3 meV is well reproduced with these values. Further, selecting the propagation angles $(0, 0)$ (blue line in Fig. 4(a)], we find that our continuum absorption has the same shape and intensity as in the experiment with a high accuracy. By changing the propagation angle, we can tune the intensity of the 1s resonance without drastically changing the shape of the continuum absorption. With a rather small angle of $\theta = \pi/9$ and the optimal $\varphi = \pi/4$ (red line), we find an extremely good agreement to the spectrum of Ref. [15].

We also compare our results against other experiments from Refs. [14] and [21] in Fig. 4(b). Again, we show $(\pi/9, \pi/4)$ with $\epsilon(w) = 47.8$ (red line) and additionally $(\pi/2, \pi/4)$ with $\epsilon(w) = 47.8$ (blue line). Once more, the excitonic resonance is reproduced with great precision. The continuum absorption of both sets settles in-between the ones from Ref. [14] (black line) and Ref. [21] (shaded area). As previously, the quadrupole interaction of the $(\pi/2, \pi/4)$ produces the highest intensity for the 1s resonance that is now roughly seven times stronger than in the experiments.

In Fig. 4 the band gaps for experimental data have been fixed so that the spectral position of the excitonic resonance matches with the result of Ref. [15], where the resonance is located at $\hbar \omega = 3.03$ eV. For our theoretical results, we do essentially the same by using the scissor shifted $E_g = 3.034$ eV. In results of Figs. 3 and 4 we have selected the scattering function $\gamma_\nu(\omega)$ in Eq. (18) to be $\gamma_{1s}(\omega) = 0.8$ meV for our 1s states, producing a similar shape and linewidth broadening for the modeled 1s resonance as seen for the experimental resonance. For all the other excitonic states, we use $\gamma_\nu = 1.8$ meV and $\mu_\nu = \Delta E_\nu = 1.2$ meV. With these selections we mimic the excitation-induced dephasing [28, 44], without drastically altering location and intensities of our resonances compared to a case $\gamma_\nu(\omega) = \gamma_\nu$, but reproduce the experimentally detected [62] steep decay of the continuum absorption.

![FIG. 4. (Color online) Experiments vs. modeled absorption. (a) Computed absorption spectrum with $\epsilon(w) = 24.3 \equiv \epsilon_1$ (black line, scaled down by dividing with 10) has a drastically different general shape than what is observed in the experiments of Ref. [15] (shaded area) and its intensity is one order of magnitude too strong. With $\epsilon(w) = 47.8 = \epsilon_2$ and light propagation parallel to $z$ axis (blue line), the shape and magnitude of the continuum absorption corresponds to the experiment. Changing the propagation to $\theta = \pi/9$ and $\varphi = \pi/4$ (red line), we can achieve similar excitonic feature as in the experiment via 1s exciton. (b) Spectrum computed with $\epsilon_2$, $\theta = \pi/9$, and $\varphi = \pi/4$ (red line) also agrees well with other experiments in Ref. [14] (black line) and in Ref. [21] (shaded area). With the angle that supports the most strongest quadrupole interaction (blue line), we find a clear 1s-exciton resonance. (inset) The 1s and 2p$_{2x}$ binding energies as a function of $\epsilon$.](Image)
absorption tail in rutile TiO$_2$.

V. SUMMARY

In summary, we systematically combine DFT and the cluster-expansion approach to compute the excitonic optical properties of semiconductors. More specifically, the matrix elements needed for microscopic modeling are computed via DFT while the many-body problem is solved with the cluster expansion. We apply this hybrid approach for rutile TiO$_2$ and compute its near-bandgap optical absorption, following from the electric-dipole, electric-quadrupole and magnetic-dipole light–matter interactions. Our results show that the quadrupole interaction in rutile is highly dependent on the propagation and polarization direction of light. Furthermore, we find that it is hard to explain the experimentally detected excitonic signature in the absorption spectrum of TiO$_2$ by considering only electric-dipole interaction and the dipole-allowed $2p$ exciton. We obtain an excellent agreement between modeled and experimental spectra through the crystallographic axes of TiO$_2$. Hence, the hybrid CE-DFT method seems to be a very promising approach to model many-body effects in semiconductors, opening a wide range of new possibilities to study and utilize properties of nontrivial systems.

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Appendix A: Renormalized terms

The exact form of the renormalized kinetic energy $\tilde{E}^{\lambda}_{ij}$ and the renormalized Rabi frequency $\Omega^{\lambda}_{ij}$ appearing in Eq. (11) are given by

\[
\tilde{E}^{\lambda}_{ij} = \delta_{ij} E_i^\lambda - E(t) D^{\lambda\lambda}_{ji} + \sum_{k,l} \left[ v_{jk;i}^{\lambda\lambda} P_{kl}^{\lambda\lambda} + v_{jk;li}^{\lambda\lambda} P_{kl}^{\lambda\lambda} \right]
+ v_{jk;li}^{\lambda\lambda} P_{kl}^{\lambda\lambda} - v^{E_i;\lambda}_{ji} , \quad (A1)
\]

\[
\Omega^{\lambda}_{ij} = E(t) D^{\lambda\lambda}_{ji} - \sum_{k,l} \left[ v_{jk;i}^{\lambda\lambda} P_{kl}^{\lambda\lambda} + v_{jk;li}^{\lambda\lambda} P_{kl}^{\lambda\lambda} \right]
+ v_{jk;li}^{\lambda\lambda} P_{kl}^{\lambda\lambda} + v^{\Omega_i;\lambda}_{ji} , \quad (A2)
\]

and have been added to ensure that the kinetic energy $A1$ and Rabi frequency $A2$ of the ground state will not become renormalized. By adding the corrections $V^{E_i;\lambda}_{ji}$ and $V^{\Omega_i;\lambda}_{ji}$ into Eqs. (A1) and (A2), we avoid double counting contributions that are already included in the effective single-particle potential $U(r)$ in Eq. (4).

As we evaluate Eq. (11) for the polarization $P_{ij}$, we find that all the electron–electron interaction terms of the form $V_{jk;li}^{\lambda\lambda\lambda\lambda}$ in Eqs. (A1) and (A2) as well as $E(t) D^{\lambda\lambda}_{ji}$ in Eq. (A1) are related to a nonlinear response. Consequently, these terms do not contribute to the linear absorption. Neglecting the $V_{jk;li}^{\lambda\lambda\lambda\lambda}$ term in Eq. (A2) that couples $P^{\lambda\lambda}_{ij}$ and $P^{\lambda\lambda}_{cd}$ polarizations is known as the Tamm-Dancoff approximation $A3. By setting $f^{ij}_{ij} = f^{ji}_{ij} = 0$, we obtain the final form of Eq. (12).

These neglected $V$ terms originate from processes where electron–electron interaction scatters electronic states between valence- and conduction-band states. When they are separated by a sufficiently large energy gap, this scattering is energetically highly unfavorable. Hence, all terms of the form $V^{\lambda\lambda\lambda\lambda}_{jk;li}$, $V^{\lambda\lambda\lambda\lambda}_{jk;li}$, and $V^{\lambda\lambda\lambda\lambda}_{jk;li}$ in Eqs. (A1) and (A2) can typically be omitted, which drastically simplifies the problem and reduces numerical efforts.

For crystalline solids (with sufficient $E_g$), these terms remain negligible as long as the system is excited in a sufficiently small region of the Brillouin zone (see Appendix C). However, the effect of these terms should be verified based on the studied properties of any new system, as we do in Appendix C for the near-bandgap optical properties of rutile TiO$_2$. For example, it is found that the Tamm-Dancoff approximation can fail in molecular systems $A4$ while in some cases it can improve the results $A5$.

Appendix B: Light–matter matrix elements

In an infinite crystal, the $r_i$-matrix element of the $i$-th spatial component of $\mathbf{r}$ becomes ambiguously defined via normal functions, but can be expressed via distributions like the Dirac delta function and its derivatives $A6, A7$. As we show later, the same holds for products of $r_i$ operators. In the scope of this work, we can properly understand these generalized functions by considering the oscillator strength, Eq. (17), that is given in a two-band model by an integration over the first Brillouin zone

\[
\mathcal{F}_\nu = \int_{BZ} d^3k \delta^3k^\nu \phi^*_\nu(k^\nu) \mathcal{F}_\nu(k^\nu), \quad (B1)
\]
and the connection between bands $\lambda = c$ and $\lambda = v$ to the wave vectors $k^c$ and $k^v$ is explicitly denoted. When the excitation of the system remains inside a sufficiently small region of the Brillouin zone, the integration boundaries can be extended to infinity since the excitonic wave functions approach zero. Assuming a parabolic energy dispersion \cite{22}, we can introduce a change of variables in Eq. \ref{B11} to the center-of-mass, $q$, and relative-momentum, $h$, wave vectors that are related to the wave vectors $k^c$ and $k^v$ by the transformation

$$k_i^c = k_i + \frac{m_e}{M_i} q_i, \quad k_i^v = k_i - \frac{m_i^h}{M_i} q_i.$$ \hspace{1cm} (B2)

Equation \ref{B11} then reads

$$F_{\nu} = \int d^3q \ d^3k \ \phi_{\nu,q}^*(k) \mathcal{R}_{k^c,k^v}^c.$$ \hspace{1cm} (B3)

We can give all terms in Eq. \ref{5} through operators $\prod_{i=1}^3 \tilde{r}_i^{n_i}$ and $\prod_{j=1}^3 r_j^{n_j}$ where $n_i$ is a non-negative integer. The matrix elements for these operators are given by

$$\mathcal{R}_{k^c,k^v}^c \equiv \int d^3r \ \psi_{c,k^c}^c(r) \left( \prod_{i=1}^3 r_i^{n_i} \right) \psi_{v,k^v}(r)$$

$$= \sum_G U_{k^c,k^v}^c(G) \frac{1}{(2\pi)^3} \int d^3r \ e^{-i(q-G)r} \left( \prod_{i=1}^3 r_i^{n_i} \right) \delta(q),$$ \hspace{1cm} (B4)

$$\mathcal{P}_{k^c,k^v}^{n_j} \equiv \int d^3r \ \psi_{c,k^c}^c(r) \left( \prod_{i=1}^3 r_i^{n_i} \right) p_j \psi_{v,k^v}(r)$$

$$= \left\langle u_{c,k^c} | u_{v,k^v} \right\rangle \Omega_0 \left( \prod_{i=1}^3 i^{n_i} \frac{\partial^{n_i}}{\partial q_i^{n_i}} \right) \delta(q)$$

$$+ \hbar \left( k_j - \frac{m_i^h}{M_j} q_j \right) \mathcal{R}_{k^c,k^v}^n,$$ \hspace{1cm} (B5)

where $n = (n_1, n_2, n_3)$ defines the integers $n_i$, and we have assumed that the excitation remains inside a region where $|q| < |G|$ for any nonzero reciprocal-lattice vector $G$. We also use the unit-cell Fourier transforms for $u_{c,k^c}^c(r) u_{v,k^v}(r)$ and $u_{c,k^c}^c(r) p_j u_{v,k^v}(r)$ [see Eqs. \ref{C2} and \ref{C3}] and denoted the matrix element of an operator $\hat{O}$

$$\left\langle u_{\lambda,k} \hat{O} u_{\nu,k'} \right\rangle = \frac{1}{\Omega_0} \int d^3r \ u_{\lambda,k}^* \hat{O} \ u_{\nu,k'}(r)$$ \hspace{1cm} (B6)

The oscillator strength \cite{B3} is related to the matrix elements in Eqs. \ref{B9} and \ref{B10} via

$$F_{\nu} \equiv \int d^3q \ d^3k \ \phi_{\nu,q}^*(k) \mathcal{R}_{k^c,k^v}^c = \int d^3k \ \phi_{\nu,0}^*(k) \mathcal{R}_{k^c}^c,$$ \hspace{1cm} (B7)

$$F_{\nu}^{n_j} = \int d^3q \ d^3k \ \phi_{\nu,q}^*(k) \mathcal{P}_{k^c,k^v}^{n_j} = \int d^3k \ \phi_{\nu,0}^*(k) \mathcal{R}_{k^c}^{n_j},$$ \hspace{1cm} (B8)

where

$$\mathcal{R}_{k^c}^n \equiv \left( \prod_{i=1}^3 (-i)^{n_i} \frac{\partial^{n_i}}{\partial q_i^{n_i}} \right) \left\langle u_{c,k^c} | u_{v,k^v} \right\rangle \Omega_0 \bigg|_{q=0},$$ \hspace{1cm} (B9)

$$\mathcal{P}_{k^c}^{n_j} \equiv \left( \prod_{i=1}^3 (-i)^{n_i} \frac{\partial^{n_i}}{\partial q_i^{n_i}} \right) \left\langle u_{c,k^c} | p_j | u_{v,k^v} \right\rangle \Omega_0 \bigg|_{q=0},$$ \hspace{1cm} (B10)

$$\mathcal{R}_{k^c}^{n_j} \equiv \left( \prod_{i=1}^3 (-i)^{n_i} \frac{\partial^{n_i}}{\partial q_i^{n_i}} \right) \left\langle u_{c,k^c} | u_{v,k^v} \right\rangle \Omega_0 \bigg|_{q=0}.$$ \hspace{1cm} (B11)

In general, a Bloch function $u_{\lambda,k}(r)$ is an analytic function of $k$ if the band $\lambda$ is not degenerate \cite{63,65}, yielding a well-defined differentiation in Eqs. \ref{B9}-\ref{B11}. A comparison between Eqs. \ref{B7}, \ref{B8}, and \ref{B11} reveals that the matrix element in Eqs. \ref{B4} and \ref{B5} are effectively direct in $\hbar$ space, i.e., their contribution to the light–matter interaction are given by

$$\mathcal{R}_{k^c,k^v}^n = \delta(k^c - k^v) \mathcal{R}_{k^c}^n,$$ \hspace{1cm} (B12)

$$\mathcal{P}_{k^c,k^v}^{n_j} = \delta(k^c - k^v) \mathcal{P}_{k^c}^{n_j} + \hbar k_j \mathcal{R}_{k^c}^{n_j} - \frac{m_i^h}{M_j} \tilde{R}_{k^c,k^v}^{n_j},$$ \hspace{1cm} (B13)

and can be determined after Eqs. \ref{B9}-\ref{B11} have been solved. For nondegenerate bands $c$ and $v$, we can do this systematically by applying $k \cdot p$ theory and the conventional $n^h$-order nondegenerate perturbation theory by expanding the functions $u_{c,k^c}$ and $u_{v,k^v}$ at $k$ where $n$ is the largest $n_i$ within Eqs. \ref{B9}-\ref{B11}.

In this work, we consider the electric-dipole, electric-quadrupole, and magnetic-dipole interactions in Eqs. \ref{24}–\ref{26} that are described by

$$D = - e \sum_i e_i r_i,$$ \hspace{1cm} (B14)

$$Q = - \frac{e^2}{2} \sum_{i,j} e_i q_j r_i r_j,$$ \hspace{1cm} (B15)

$$M = - e \frac{2 \omega m_0}{\lambda} \sum_{i,j} (e_i q_j - e_j q_i) r_i p_j.$$ \hspace{1cm} (B16)

We obtain the matrix elements in Eqs. \ref{27}–\ref{29} by using Eqs. \ref{B9}-\ref{B13}, $k \cdot p$ theory, and the second order perturbation theory for matrix elements of operators in Eqs. \ref{B14}-\ref{B16}. This rather lengthy but straightforward calculation yields $Q_{k;i;j}^{cv} = Q_{k;i;j}^{cv} + \tilde{Q}_{k;i;j}$ and

The oscillator strength \cite{B3} is related to the matrix.
\[ M_{kij}^{cv} = M_{kij} - \tilde{M}_{kij}^{cv} \]
\[ \tilde{Q}_{kij}^{cv} = -\frac{i}{2} \left( 1 - \frac{m_f^2 m_h^2}{M_f M_h} \right) q_{kij}^{cv} \]
\[ + \frac{e^2}{2 m_0} \left[ M_i M_j (E_k^c - E_{k_i}^c) \right]^{-1} \left[ m_i^b m_j^b \right] \left[ s_{kij}^{cv} \right] \]
\[ - p_{k}^c p_{k}^{c*} \left[ \left( q_{kij}^{cv} \right) - p_{k}^{c} p_{k}^{c*} \right] \]
= \frac{e}{2 m_0} \left[ h k_j P_{k+i}^c + \frac{m_f^2}{M_f} s_{kij}^{cv} - \frac{m_h^2}{M_h} \left( s_{kij}^{cv} \right) \right]. \]

Here, we have defined
\[ q_{kij}^{\lambda \lambda'} = \sum_{\eta \neq \lambda, \lambda'} \tilde{p}_{kij}^{\lambda \eta} \rho_{kij}^{\lambda \lambda'}, \quad s_{kij}^{\lambda \lambda'} = \sum_{\eta \neq \lambda} \tilde{p}_{kij}^{\lambda \eta} \rho_{kij}^{\lambda \lambda'} \]
where \( \tilde{p}_{kij}^{\lambda \eta} \) (\( \rho_{kij}^{\lambda \lambda'} \)) is the \( i \)th component of the vector \( \tilde{p}_{kij}^{\lambda \lambda'} \) (\( \rho_{kij}^{\lambda \lambda'} \)).

**Appendix C: Coulomb interaction in crystals**

In a crystal with electronic wave functions \([21]\) and electron–electron interaction \([22, 23]\), the explicit Coulomb matrix element in Eq. (9) becomes
\[ V_{k_1 k_2 k_3 k_4}^{\lambda_1 \lambda_2 \lambda_3 \lambda_4} = \sum_{G, G'} U_{k_1 k_4}^{\lambda_1 \lambda_4} (G) U_{k_2 k_3}^{\lambda_2 \lambda_3} (G') V_{k_3 - k_2 + G'} \]
\[ \times \delta (G + G' - k_1 - k_2 + k_4 + k_3), \]
(C1)

where we have expressed the Fourier transform of pairs of Bloch functions as a sum
\[ u_{\lambda, k}^{\gamma}(r) u_{\lambda', k'}^{\gamma}(r) = \sum_{G} U_{kk'}^{\lambda \lambda'} (G) e^{i G \cdot r}, \]
(C2)
\[ U_{kk'}^{\lambda \lambda'} (G) = \frac{1}{V_0} \int_{V_0} d^3 r u_{\lambda, k}^{\gamma}(r) u_{\lambda', k'}^{\gamma}(r) e^{-i G \cdot r}, \]
(C3)

over the reciprocal lattice vectors \( G \). If we now consider a system where the excitation region inside a region where \( |k_1 + k_2 - k_4 - k_3| < |G| \) for any nonzero \( G \), Eq. (C1) produces
\[ \tilde{V}_{kk'q}^{\lambda_1 \lambda_2 \lambda_3 \lambda_4} = \delta (k_1 + k_2 - k_3 - k_4) \tilde{V}_{kk'q}^{\lambda_1 \lambda_2 \lambda_3 \lambda_4} \]
\[ \times \sum_{G \neq 0} \tilde{U}_{k'k}^{\lambda_3 \lambda_4} (G) \]
\[ \times \tilde{U}_{k'k}^{\lambda_1 \lambda_2 \lambda_3 \lambda_4} (G) \]
\[ \times \tilde{U}_{k'k}^{\lambda_1 \lambda_2 \lambda_3 \lambda_4} (G) \]
\[ \times \tilde{U}_{k'k}^{\lambda_1 \lambda_2 \lambda_3 \lambda_4} (G) \]
\[ \times \tilde{U}_{k'k}^{\lambda_1 \lambda_2 \lambda_3 \lambda_4} (G) \]
contains the matrix element
\[ \tilde{V}_{kk'q}^{\lambda_1 \lambda_2 \lambda_3 \lambda_4} = V_q (u_{\lambda_1, k+q} | u_{\lambda_4, k} \rangle \langle u_{\lambda_2, k'} | u_{\lambda_3, k'} \rangle \langle 0_0 |. \]
(C6)

Since \( \tilde{V}_{kk'q}^{\lambda_1 \lambda_2 \lambda_3 \lambda_4} \) diverges at \( q = 0 \), the leading Coulomb-interaction contributions will follow from this term. The actual \( q = 0 \) divergence can be removed by using the jellium model \([2]\). We next consider excitations where \( \langle q | | G \rangle \) for any nonzero \( G \) so that we can approximate \( \langle u_{\lambda, k} | u_{\lambda', k+q} \rangle \langle 0_0 | = \delta_{\lambda \lambda'} \). If we only allow contributions that are proportional to \( 1/|q| \), the matrix element \([44]\) directly follows from \([C5]\) (see Ref. \([24]\) for a similar approach).

We next check the approximated Coulomb matrix elements for TiO\(_2\). To do this, we consider the most important direct elements given by \( V_{kk'q}^{\lambda_1 \lambda_2 \lambda_3 \lambda_4} \). For these terms we study how well the approximate Coulomb matrix element \( V_q \) compares with the explicit \( V_{kk'q}^{\lambda_1 \lambda_2 \lambda_3 \lambda_4} \) obtained from DFT. By performing a great number of control computations using \( k \) grids with \( |k| a_0 < 10 \) and \( |q| a_0 < 10 \), we find a maximal error of only 3\% while in the most important region for the convergence of excitonic states the error is less than 10\(^{-3}\).

We also have computed the matrix elements \( V_{kk'q}^{\lambda_1 \lambda_2 \lambda_3 \lambda_4} \) and \( V_{kk'q}^{\lambda_1 \lambda_2 \lambda_3 \lambda_4} \) that contribute to the linear response if the Tamm-Dancoff approximation is not made and the exchange terms in Eqs. \([12, 13]\) are not omitted. In the region of interest, these Coulomb contributions are negligible compared to the leading Coulomb terms and orders of magnitude smaller than \( E_g \). Hence, it is justified to omit these terms.

**Appendix D: Excitonic states**

Excitonic states are defined by the Wannier equation \([14]\). Using the effective mass approximation Eq. \([21]\), the Coulomb matrix elements Eq. \([8]\), and the two dielectric constants of TiO\(_2\), it is given by
\[ \left\{ \frac{\hbar^2}{2 \mu_{\perp}} k_1^2 + \frac{\hbar^2}{2 \mu_{\parallel}} k_2^2 \right\} \phi_{\lambda}(k) - \frac{e^2}{4 \pi \varepsilon_0 \varepsilon_{\perp}} \int d^3 k' \phi'_{\lambda}(k') \]
\[ \times \left[ (k_{\perp} - k_{\perp}')^2 + \frac{\varepsilon_{\parallel}}{\varepsilon_{\perp}} (k_{\parallel} - k_{\parallel}')^2 \right]^{-1} = E_{\lambda} \phi_{\lambda}(k) , \]
(D1)

where the momentum \( k \equiv (k_{\perp}, k_{\parallel}) \) has been decomposed into directions in \( (k_{\perp}) \) and out-of \( (k_{\parallel}) \) the \( k_{\perp} \) plane. It is beneficial to make the coordinate transformation \( k_{\perp} \rightarrow \sqrt{\varepsilon_{\perp}/\varepsilon_{\parallel}} k_{\perp} \) that converts Eq. \([D1]\) into
\[ \left\{ \frac{\hbar^2}{2 \mu_{\perp}} k_1^2 + \frac{\hbar^2}{2 \mu_{\parallel}} k_2^2 \right\} \phi'_{\lambda}(k) \]
\[ - \frac{e^2}{4 \pi \varepsilon_0 \varepsilon' } \int d^3 k' \phi'_{\lambda}(k') \]
\[ = E_{\lambda} \phi'_{\lambda}(k), \]
(D2)

where \( \mu_{\parallel} = \mu_{\parallel} / \varepsilon_{\parallel}, \quad \varepsilon' = \sqrt{1 + \varepsilon_{\perp}}, \) and \( \phi'_{\lambda}(k) = \phi_{\lambda}(k_{\perp}, \sqrt{\varepsilon_{\parallel}/\varepsilon_{\perp}} k_{\parallel}) \).

To solve Eq. \([D2]\) efficiently, we expand the wave functions \( \phi'_{\lambda}(k) \) using spherical harmonics \( Y_{lm}(\theta, \varphi) \)
\[ \phi'_{\lambda}(k) = \sum_{l, m} R_{\lambda, l, m}(k) Y_{lm}(\theta, \varphi), \]
(D3)
where \( l = 0, 1, 2, \ldots \) and \( |m| < l \) are the angular and magnetic quantum numbers, respectively, and \( R_{\lambda,m}(k) \) is the radial component. Projecting Eq. (D2) to spherical harmonics using the ansatz \( \langle \ell m | \ell^l \rangle \) yields an eigenvalue problem for the radial part alone that can be solved numerically. A detailed description of this method will be published elsewhere \[54\]. The expansion Eq. (D3) converges fast in terms of angular quantum numbers. To obtain the required wave functions, we included five \( l \) states.

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