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Rukelj, Zoran; Despoja, Vito

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PAPER

Estimation of the single-particle band gap and exciton binding energy in two dimensional insulators: a modified G_0W_0 -BSE method approachZoran Rukelj¹ and Vito Despoja² ¹ Department of Physics, University of Zagreb, Bijenička 32, HR-10000 Zagreb, Croatia² Institute of Physics, Bijenička 46, HR-10000 Zagreb, CroatiaE-mail: vito@phy.hrKeywords: excitons, 2D crystals, optical properties, G_0W_0 band-gap

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Abstract

In this paper we present an alternative G_0W_0 -BSE procedure, suitable for calculation of the quasi-particle and optical properties in 2D semiconductors. The method completely excludes the spurious Coulomb interaction with 2D crystal replicas. The calculated band gap energies of hexagonal boron nitride (hBN), MoS_2 and MoTe_2 monolayers are in good agreement with other theoretical results. The 2D Bethe–Salpeter equation is derived and reduced to a 2D-hydrogen Schrödinger equation in which enter the G_0W_0 band gap, DFT effective masses, and RPA screened Coulomb interaction. This formulation is applied to the problems of determining exciton binding energies and estimating the quasiparticle band gap in hBN, as well as in some transition-metal dichalcogenides. A semiclassical procedure is used in the limit of high polarizability λ in order to obtain the analytical expression for exciton binding energies.

1. Introduction

For the last decade, the condensed matter physics has been dominated by the experimental and theoretical investigations of two-dimensional materials. One class of such materials are the direct gap two-dimensional insulators. The best known examples are monolayers (ML) of the members of transition-metal dichalcogenides (TMDs) [1, 2] and the hexagonal boron nitride (hBN) monolayer. Many interesting phenomena are attributed to TMDs and hBN. Heterostructures with hBN improve graphene transport properties, enhancing its application in electronics [3, 4]. By applying uniaxial or biaxial strain, the optical properties of single-layer TMDs can be tuned [5, 6]. Single-layer TMD (MoS_2 or WS_2) transistors are synthesized and tested for application in digital electronics [7–9]. Finally, TMDs can be potentially applied in optoelectronics as photodetectors [10] or as photodiodes and photovoltaic devices [11–13].

Although hBN and TMDs appear as simple honeycomb crystals with just two or three atoms per unit cell, theoretical calculations of their electronic structures and optical properties tend to be complex. Specifically, the very inefficient screening causes the long-range Coulomb correlations to prevail and the DFT methodology systematically fails in estimating the single particle band gap. In addition, the strong (weakly screened) Coulomb interaction gives rise to excitons with large binding energies.

The well-established *ab initio* methodology which includes quasiparticle corrections (GW method) and solving the Bethe–Salpeter equation (BSE, the so-called GW-BSE method) is capable of giving a very accurate single particle band gap, excitonic binding energies, and oscillator strengths [14–20]. The semi-analytic study of optical properties in hBN/substrate systems in the framework of Bethe–Salpeter and Wannier–Keldysh approach were done in reference [21]. The authors provided very extensive investigation of excitons, exciton-polaritons, absorption and spontaneous radiative decay of excitons in hBN on SiO_2 or graphite surfaces. The authors also presented the analytic expressions for hBN/surface absorption spectra and quasi-particle corrections of hBN/surface band gaps. They suggested how their semi-analytic approach

can be generalised to various TMDs. Moreover, in reference [22] the authors study how the proximity effects, due to the presence of a ferromagnetic cobalt thin film, modify the excitonic properties and optical conductivity in MoS₂.

The extended calculations of the quasiparticle and exciton properties in hBN-ML followed by the *ab initio* parameterized TBA are also provided [23]. However, the GW-BSE method is computationally heavy and time consuming. Therefore, it is useful to have simpler complementary numerical tools that allow a faster estimation of band gaps, as well as exciton binding energies.

In this paper we introduce an alternative G_0W_0 method adapted for the calculation of quasi two dimensional (q2D) quasiparticle properties, the q2D- G_0W_0 method. The primary advantage of q2D- G_0W_0 method is that it completely excludes the spurious Coulomb interaction with 2D crystal replicas. The computational acceleration is achieved through the inclusion of the crystal-local-field-effects (CLFE) in the correlation self-energy $\Sigma_{n\mathbf{k}}^C(\omega)$ solely in the perpendicular direction. The obtained q2D- G_0W_0 band-gaps in hBN, MoS₂ and MoTe₂ are in good agreement with other theoretical results [20, 24–27]. The exciton binding energies in hBN will be explored by solving a computationally much simpler 2D Bethe-Salpeter equation (2D-BSE) equation which is derived from the 2D equation of motion for the electron–hole pair propagator [28, 29]. The q2D- G_0W_0 band-gap (through the definition of screening length) and RPA 2D screened Coulomb interaction w enter into 2D-BSE. The plausibility of 2D-BSE will be commented upon. In the approach of two parabolic bands (described by effective masses m_v^* and m_c^*) and in the long wavelength limit ($Q \rightarrow 0$) the 2D-BSE is reduced to a two body Schrödinger equation in a screened DFT 2D Coulomb potential w . A realistic value of the single particle band gap will be obtained using the derived formulas and the experimentally determined exciton energy. In particular, the ground state exciton binding energy in the high polarizability limit will be investigated by employing the Einstein–Brillouin–Keller (EBK) procedure [30] and compared with the solutions of the Schrödinger equation. The analytical expression for the ground state exciton binding energy will be derived and used to predict the quasiparticle band gaps of the selected two-dimensional insulators [18–20].

This paper is organized as follows: in section 2 the formulation of the q2D- G_0W_0 method is presented and applied to calculate the hBN band gap in G , K and M points of Brillouin zone. Then, the 2D-BSE is derived from equations of motion for the electron–hole propagator with the implemented RPA 2D screened Coulomb interaction w . The 2D-BSE is reduced to a two-body Schrödinger equation. The results are presented in section 3. The first ten exciton energy levels are calculated, as well their spatial extent. The derived formulation is used to estimate the quasiparticle band gap in hBN and various TMDs. The ground state exciton binding energy in the high polarizability limit is investigated using the EBK procedure and compared with the results of section 3.1. Conclusions are presented in section 4.

2. *Ab initio* studies of hBN-ML quasi-particle and optical properties

In this section the modified q2D- G_0W_0 method adapted for the calculation of quasiparticle band gaps in 2D crystals is presented. The equation of motion is formulated for the 2D electron–hole operator which is analogous to 2D-BSE. This equation contains q2D- G_0W_0 corrected band gap and RPA 2D screened Coulomb interaction w . Finally, it is shown that this methodology produces a fairly accurate band-gap and exciton binding energies.

2.1. The q2D- G_0W_0 method

It is well known that the LDA underestimates the semiconducting band gap. Therefore, in order to obtain the accurate exciton energy, providing quasiparticle corrections is of crucial importance. The calculation of the exchange–correlation self-energy within G_0W_0 approximation, introduced a long time ago by Hedin, Strinati and Louie [31–34] and more recently by Louie, Thygesen and others [20, 24, 35], is applied to calculate various crystal quasiparticle band structures. Here we briefly present some modifications which should be done to the standard G_0W_0 approximation when used in calculating the band gaps of 2D crystals. The G_0W_0 exchange–correlation (XC) self-energy corresponding to the Bloch state (n, \mathbf{k}) is

$$\Sigma_{n\mathbf{k}}^{\text{XC}}(\omega) = \Sigma_{n\mathbf{k}}^{\text{X}} + \Sigma_{n\mathbf{k}}^{\text{C}}(\omega), \quad (2.1)$$

where the exchange self-energy is

$$\Sigma_{n\mathbf{k}}^{\text{X}} = -\frac{1}{V} \sum_m \sum_{\mathbf{G}\mathbf{G}' \mathbf{q} \in B.Z} f_{\mathbf{k}+\mathbf{q}}^m \rho_{n\mathbf{k},m\mathbf{k}+\mathbf{q}}^*(\mathbf{G}) v_{\mathbf{G}\mathbf{G}'}(\mathbf{q}) \rho_{n\mathbf{k},m\mathbf{k}+\mathbf{q}}(\mathbf{G}') \quad (2.2)$$

and the correlation self-energy is

$$\Sigma_{nk}^C(\omega) = \frac{1}{V} \sum_m \sum_{\mathbf{G}\mathbf{G}'} \sum_{\mathbf{q} \in B.Z} \rho_{nk, mk+\mathbf{q}}^*(\mathbf{G}) \left[i \int_{-\infty}^{\infty} \frac{d\nu}{2\pi} e^{i\nu\tau} G_{mk+\mathbf{q}}^0(\omega + \nu) W_{\mathbf{G}\mathbf{G}'}^{\text{ind}}(\mathbf{q}, \nu) \right] \rho_{nk, mk+\mathbf{q}}(\mathbf{G}'). \quad (2.3)$$

Here $f_{\mathbf{k}}^n$ is the Fermi–Dirac distribution, $G_{nk}^0(\omega) = 1/[\hbar\omega - E_{\mathbf{k}}^n + i\eta \text{sgn}(E_{\mathbf{k}}^n - E_F)]$ is the one-particle Green’s function at $T = 0$ and the charge vertices are

$$\rho_{nk, mk+\mathbf{q}}(\mathbf{G}) = \int_V d\mathbf{r} \phi_{nk}^*(\mathbf{r}) e^{-i(\mathbf{q}+\mathbf{G})\mathbf{r}} \phi_{mk+\mathbf{q}}(\mathbf{r}). \quad (2.4)$$

The $\tau = 0^+$ and $\eta = 0^+$ are infinitesimally small positive numbers. The induced part of dynamically screened Coulomb interaction (which enters into (2.3)) can be written as

$$W_{\mathbf{G}\mathbf{G}'}^{\text{ind}}(\mathbf{Q}, \omega) = \sum_{\mathbf{G}_1 \mathbf{G}_2} v_{\mathbf{G}\mathbf{G}_1}(\mathbf{Q}) \chi_{\mathbf{G}_1 \mathbf{G}_2}^T(\mathbf{Q}, \omega) v_{\mathbf{G}_2 \mathbf{G}'}(\mathbf{Q}). \quad (2.5)$$

The time-ordered response function is the solution of the Dyson equation

$$\chi_{\mathbf{G}\mathbf{G}'}^T(\mathbf{q}, \omega) = \tilde{\chi}_{\mathbf{G}\mathbf{G}'}^0(\mathbf{q}, \omega) + \sum_{\mathbf{G}_1 \mathbf{G}_2} \tilde{\chi}_{\mathbf{G}\mathbf{G}_1}^0(\mathbf{q}, \omega) v_{\mathbf{G}_1 \mathbf{G}_2}(\mathbf{q}) \chi_{\mathbf{G}_2 \mathbf{G}'}^T(\mathbf{q}, \omega), \quad (2.6)$$

where the RPA time-ordered irreducible polarizability is explicitly given by

$$\tilde{\chi}_{\mathbf{G}\mathbf{G}'}^0(\mathbf{q}, \omega) = \frac{2}{V} \sum_{nm} \sum_{\mathbf{k} \in B.Z} \rho_{nk, mk+\mathbf{q}}(\mathbf{G}) \rho_{nk, mk+\mathbf{q}}^*(\mathbf{G}') \frac{f_{\mathbf{k}}^n - f_{\mathbf{k}+\mathbf{q}}^m}{\hbar\omega + E_{\mathbf{k}}^n - E_{\mathbf{k}+\mathbf{q}}^m + i\eta \text{sgn}(E_{\mathbf{k}+\mathbf{q}}^m - E_{\mathbf{k}}^n)}. \quad (2.7)$$

The Coulomb interaction kernel is

$$v_{\mathbf{G}\mathbf{G}'}(\mathbf{Q}) \equiv v_{\mathbf{G}\mathbf{G}'}^{3D}(\mathbf{Q}) = \frac{4\pi}{|\mathbf{Q} + \mathbf{G}|^2} \delta_{\mathbf{G}\mathbf{G}'}, \quad (2.8)$$

$\mathbf{q} = (q_x, q_y, q_z)$ is a 3D momentum transfer, $\mathbf{r} = (\rho, z)$ is a 3D position vector, $\mathbf{G} = (\mathbf{G}_{\parallel}, G_z)$ are the reciprocal lattice wave vectors and V is the normalization volume. The equations (2.1)–(2.7) represent the standard G_0W_0 procedure for the electronic self-energy calculation in 3D crystals. The question now is how to transform these formulas in a way that they can be applied to the calculation of the electron self-energy in an isolated 2D crystal. Suppose that we want to calculate the electron self-energy in the i -th hBN monolayer which is part of periodically stacked non-overlapping hBN layers, as sketched in figure 1(a).

The charge density fluctuations $\rho_{nk, mk+\mathbf{q}}$ created by electron–electron scattering or by electron–hole transitions $|nk\rangle \rightarrow |mk+\mathbf{q}\rangle$ in i th hBN layer are represented by the blue dots while the red squares represent the same in the surrounding hBN layers ($j \neq i$). A blue dot can interact with a blue dot in the same layer via Coulomb interaction (2.8), as well as with red squares in the surrounding hBN layers. The corresponding correlation self-energy (2.3) is represented by the Feynman diagrams in figure 1(b). It consists of the process (I) where the electron propagator in i th layer interacts with the polarization in the same layer, and the processes (II–IV) where it interacts with the polarization in the surrounding layers. Since we are interested in how the electronic state in the i th layer is influenced only by the polarization of the same layer, the processes (II–IV) should be neglected and only the process (I) should be retained. However, the process (I) still contains the influence of the surrounding ($j \neq i$) hBN layers. The time-ordered RPA response function χ^T , represented by the Feynman diagrams in figure 2(a), consists of processes where the ‘bubble’ diagrams in the i th layer interact with the ‘bubble’ diagrams in the same layer (processes (i–iii)), but also of processes where the ‘bubble’ diagrams in the i th layer interact with the ‘bubble’ diagrams in the surrounding layers (e.g. process (iv)). Therefore the processes of type (iv) should be neglected and the Dyson equation for χ^T takes the form shown by Feynman diagrams figure 2(b). Finally, this results in the complete annulment of the interaction with the surrounding hBN layers. Technically, the processes (II–IV) in figure 1(b) and processes of type (iv) in figure 2(a) can be neglected in a way that the formulation (2.1)–(2.7) is retained, however, the Coulomb interaction kernel (2.8) is modified as follows [44, 45]

$$v_{\mathbf{G}\mathbf{G}'}^{3D}(\mathbf{Q}) \rightarrow v_{\mathbf{G}\mathbf{G}'}^{2D}(\mathbf{Q}) = v_{\mathbf{G}\mathbf{G}'}^{3D}(\mathbf{Q}) - p_{G_z} p_{G_z'} \frac{4\pi(1 - e^{-|\mathbf{Q}+\mathbf{G}_{\parallel}|L})}{|\mathbf{Q} + \mathbf{G}_{\parallel}|L} \frac{|\mathbf{Q} + \mathbf{G}_{\parallel}|^2 - G_z G_z'}{(|\mathbf{Q} + \mathbf{G}_{\parallel}|^2 + G_z^2)(|\mathbf{Q} + \mathbf{G}_{\parallel}|^2 + G_z'^2)} \delta_{\mathbf{G}_{\parallel} \mathbf{G}_{\parallel}'}, \quad (2.9)$$

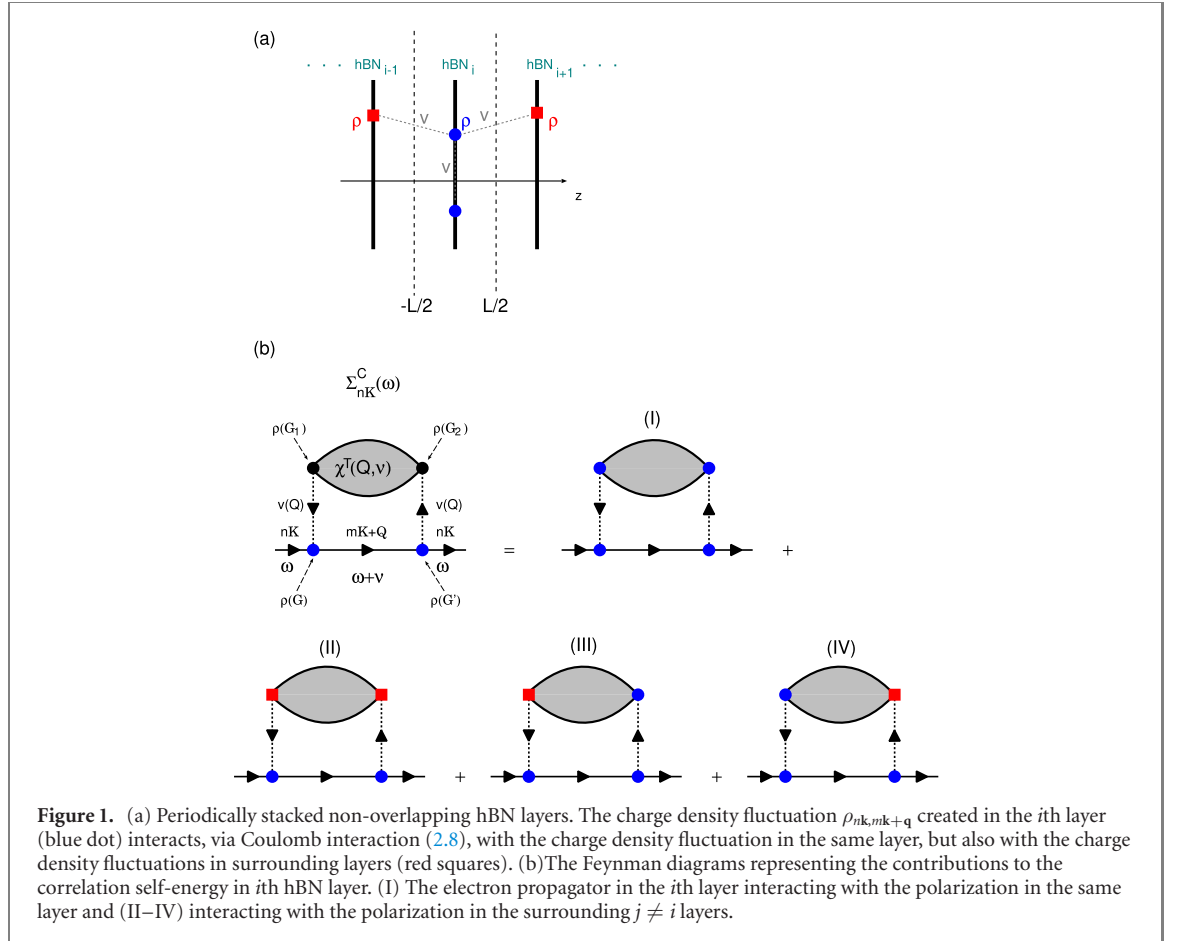


Figure 1. (a) Periodically stacked non-overlapping hBN layers. The charge density fluctuation $\rho_{nk, mk+q}$ created in the i th layer (blue dot) interacts, via Coulomb interaction (2.8), with the charge density fluctuation in the same layer, but also with the charge density fluctuations in surrounding layers (red squares). (b) The Feynman diagrams representing the contributions to the correlation self-energy in i th hBN layer. (I) The electron propagator in the i th layer interacting with the polarization in the same layer and (II–IV) interacting with the polarization in the surrounding $j \neq i$ layers.

where

$$p_{G_z} = \begin{cases} 1; & G_z = \frac{2k\pi}{L} \\ -1; & G_z = \frac{(2k+1)\pi}{L},, \quad k = 0, 1, 2, 3, \dots \end{cases}$$

Also, the following transformation must be done

$$\mathbf{q}, \mathbf{k} \rightarrow \mathbf{Q}, \mathbf{K} \quad (2.10)$$

where $\mathbf{Q} = (q_x, q_y)$ and $\mathbf{K} = (K_x, K_y)$ are the wave-vectors parallel to the x - y plane. The 3D Brillouin zone \mathbf{q} summation should be replaced by 2D summation as

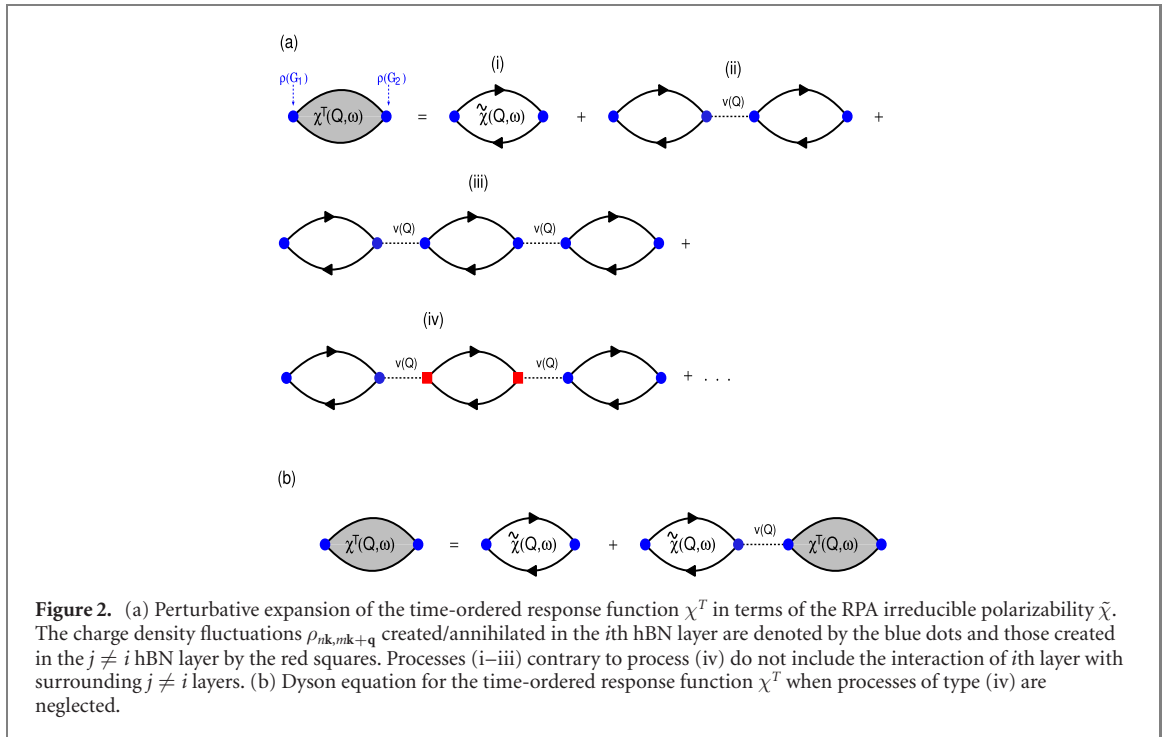
$$\frac{1}{V} \sum_{\mathbf{q} \in B.Z} \rightarrow \frac{1}{LS} \sum_{\mathbf{Q} \in S.B.Z}, \quad (2.11)$$

and the same is true for the \mathbf{k} summation in (2.7). Here S is the normalization surface and L is the unit cell constant in the perpendicular (z) direction, as sketched in figure 1(a). An example of how to transform the Dyson equation for 3D or bulk response function (2.6) into the Dyson equation for q2D response function which represents the charge density fluctuations in an isolated 2D crystal is presented in appendix A. There, it is clearly demonstrated that the only modification which should be done in (2.6) is the transformation (2.9).

The quasiparticle energy of the Bloch state (n, \mathbf{K}) is calculated by solving the equation

$$\hbar\omega + V_{n\mathbf{K}}^{XC} - \text{Re}\Sigma_{n\mathbf{K}}^{XC}(\omega) = E_{\mathbf{K}}^n, \quad (2.12)$$

for $\hbar\omega$. Here $V_{n\mathbf{K}}^{XC}$ represents the LDA exchange–correlation energy of Bloch state $|n\mathbf{K}\rangle$. For non-overlapping layers, in the exchange term (2.2) the charge density fluctuation $\rho_{nk, mk+q}$ always interacts within the same layer through the bare Coulomb interaction. This means that the expression (2.2) can still be used without further modifications.



The only difference between a Coulomb kernel (2.9) and a Coulomb truncation (equation 16) in reference [46] or (equation 43) in reference [24] is that in equation (2.9) the Coulomb interaction is ‘closed’ within $z \in [-L/2, L/2]$, i.e. the charge at $z = L/2$ cannot interact with any other charge, even at $z = L/2 + 0^+$, while the Coulomb interaction (equation (16) or (43)) is such that the charge at $z = L/2$ can interact with all of the charges in the region $z \in [L/2 - R, L/2 + R]$. This requires more layers of vacuum if one wants to avoid interaction with neighbouring cells. The detailed description how (2.9) is derived is given in reference [45]. The truncated interaction (2.9) enables that in a periodic superlattice arrangement it is sufficient to choose inter-layer spacing such that electronic densities in adjacent layers do not overlap. Any increase in this distance will not change the result. So the interaction (2.9) allows for supercells with a very few layers of vacuum, which is another important advantage over the truncated interaction proposed in reference [46].

The introduction of the Coulomb kernel (2.9) easily overcomes the most critical issue of the 3D periodicity: how to calculate the long-range image potential at any point z produced by an individual slab, located at $-L/2 < z < L/2$, which is a part of the super-lattice. This issue is also solved in reference [47]. Here, in appendix A is demonstrated that if the Dyson equation for the response function $\chi^T(z, z')$ is solved in the restricted real space $-L/2 < z, z' < L/2$, the density–density Coulomb interaction remains limited within the boundaries $[-L/2, L/2]$. This results in $\chi^T(z, z')$ representing the dynamical response of an individual slab without any admixture of other slabs. Consequently, for the induced Coulomb interaction

$$W^{\text{ind}}(Q, \omega, z, z') = \int_{-L/2}^{L/2} dz_1 dz_2 v^{2D}(Q, z, z_1) \chi^T(Q, \omega, z_1, z_2) v^{2D}(Q, z_2, z') \quad (2.13)$$

the other layers becomes ‘invisible’, i.e. it properly describes the induced interaction in point z produced by charge in z' where z and z' are no longer limited to $-L/2 < z, z' < L/2$. Here the bare Coulomb interaction $v^{2D}(Q, z, z')$ is given by (A.4). Moreover, the integration of static potential $W^{\text{ind}}(Q, \omega = 0, z, z)$ over parallel wave vector Q provides the correct image potential at any point z , as presented in detail in reference [48]. A very similar argumentation applies to calculation of the correlation self-energy (2.3) with the induced interaction (2.5), which after transformation $v \rightarrow v^{2D}$, represents exactly the Fourier transform of (2.13).

2.1.1. Implementation of the crystal local field effects in perpendicular z direction

The crystal local field effects CLFE can be explained as dispersivity (or microscopic distribution) of induced potential W^{ind} or induced density ρ^{ind} on the unit cell scale. For example, if a unit charge is placed somewhere in the crystal it will induce a charge density at macroscopic scale but also variations of microscopic charge density within the unit cell. If the CLFE are excluded then the variations of microscopic charge density are neglected and the macroscopic ones remain. From the point of view of the reciprocal

space the exclusion of CLFE results that in matrices $\tilde{\chi}^0$, χ^T and W^{ind} we retain just the $\mathbf{G} = \mathbf{G}' = 0$ component. In bulk crystals the exclusion of the CLF is sometimes justified, especially in long wavelength limit $Q \rightarrow 0$. However, in the low dimensional crystals (0D, 1D or 2D crystals) the the strong dispersivity (fast variation) of the induced charge density in the direction of the broken periodicity (e.g. in 2D crystals it would be perpendicular to crystal plane) requires inclusion of the CLFE, at least in the direction of the broken periodicity. For these reasons in this calculation we retain the CLFE in the perpendicular z direction and neglect them in the parallel direction. From the point of view of the reciprocal space this means that in all equations (2.1)–(2.9) we put $\mathbf{G}_{\parallel} = \mathbf{G}'_{\parallel} = 0$, but we keep the z component different from zero, i.e. $G_z \neq 0$ and $G'_z \neq 0$. By doing so we capture the important microscopic variations of the induced density in z direction and ignore the not so important microscopic dispersivity in parallel (x – y) directions. This reduces the dimension of matrices $\tilde{\chi}^0$, χ^T and W^{ind} , significantly reducing the memory allocation requirements and substantially accelerating the computation. At the same time the accuracy of the calculation remains satisfactory. The inclusion or exclusion of the CLFE does not affect the definition of the truncated interaction (2.9) and it remains unchanged. For example, if the CLFE are included just in z direction, then $\mathbf{G}_{\parallel} = \mathbf{G}'_{\parallel} = 0$ and the irreducible polarisability and the response function matrices become $\tilde{\chi}_{G_z G'_z}^0$ and $\chi_{G_z G'_z}^T$, respectively. This implies that in the matrix Dyson equation for χ^T (2.6) matrix v becomes $v_{G_z G'_z}^{\text{2D}}$, i.e. it has the same form as (2.9) with the exception that $\mathbf{G}_{\parallel} = \mathbf{G}'_{\parallel} = 0$.

In the optical limit ($Q \ll 1/a, 1/L$) a q2D crystal can be considered a 2D sheet crystal. In this case we can average the dynamical response in the z direction

$$\frac{1}{L} \int_{-L/2}^{L/2} dz dz' \tilde{\chi}^0(z, z') = \tilde{\chi}_{G_z=0 G'_z=0}^0,$$

and squeeze it in $z = 0$ plane

$$\tilde{\chi}_{\text{2D}}^0(z, z') = L \tilde{\chi}_{G_z=0 G'_z=0}^0 \delta(z) \delta(z').$$

In this way the strong dispersivity in z direction counterintuitively resulted in exclusion of the crystal local field effects in the z direction ($G_z = G'_z = 0$). This works well in the $Q \ll 1/a, 1/L$ limit, but for example in the correlation self energy (2.3) short wavelength contributions ($Q \approx 1/a, 1/L$) or the dispersivity of the induced interaction W^{ind} within the unit cell become important and the 2D model is no longer adequate. So when solving the BSE we use the 2D limit ($\mathbf{G} = \mathbf{G}' = 0$) but when we provide the quasiparticle GW corrections we include the CLFE in the direction in which they are strongest, the z direction, so we put $\mathbf{G}_{\parallel} = \mathbf{G}'_{\parallel} = 0$ and leave $G_z \neq 0, G'_z \neq 0$.

2.1.2. Calculation of quasi-particle band gap

At the DFT stage of the calculation the Kohn–Sham (KS) wave functions $\phi_{n\mathbf{k}}^*$ and energy levels $E_{n\mathbf{K}}$, i.e. the band structure of a hBN-ML, are determined using the plane-wave DFT code QUANTUM ESPRESSO (QE) [36]. The core–electron interaction is approximated by the norm-conserving pseudopotentials [37], and the exchange correlation (XC) potential by the LDA Perdew–Zunger (PZ) functional [38]. For the hBN-ML primitive cell constant, $a = 4.746a_0$ (a_0 is the Bohr radius) is used and the superlattice constant in the z direction is $L = 23.73a_0$. The ground state electronic densities of the hBN-ML are calculated using the $12 \times 12 \times 1$ Monkhorst–Pack \mathbf{k} -point mesh [39] of the first Brillouin zone (BZ). For the plane-wave cut-off energy we used 60 Ry (816 eV). The RPA time-ordered irreducible polarizability (2.7) is calculated using $201 \times 201 \times 1$ \mathbf{k} -point mesh sampling. This \mathbf{k} -point mesh sampling enables the minimum transfer wave vector $q_{\text{min}} = 0.0076 a_0^{-1}$. The damping parameter used is $\eta = 50$ meV and the temperature is $k_B T = 10$ meV. The band summation is performed over 40 bands, which proved sufficient for the proper description of the electronic excitations up to 30 eV. The crystal local field effects (CLFE) are included only in the perpendicular (z) direction and CLFE energy cut-off is chosen to be 10 Ry (136 eV). The \mathbf{Q} -point mesh used in self-energy calculation (2.1)–(2.7) is $21 \times 21 \times 1$. In the exchange energy calculation (2.2) the 3D CLFE are included and CLFE energy cut off is is chosen to be 20 Ry (272 eV).

The hBN-ML is a direct gap insulator with the conduction band minimum and the valence band maximum located at the K point of the Brillouin zone. The DFT band gap obtained in this calculation is $\Delta_{\text{DFT}} = 4.5$ eV and after the q2D- G_0W_0 quasiparticle correction (2.1)–(2.12) it is increased to $\Delta_{\text{GW}} = 7.7$ eV. The value of the q2D- G_0W_0 band gap is slightly overestimated compared to the value of about 7.4 eV proposed in previous calculations [23, 24, 40, 41]. The slight difference may arise from the fact that in the correlation self-energy (2.3) the CLFE are included only in the z direction. On the other hand, it seems that the number of 40 bands is well converged, considering that after using 80 bands the gap was reduced by

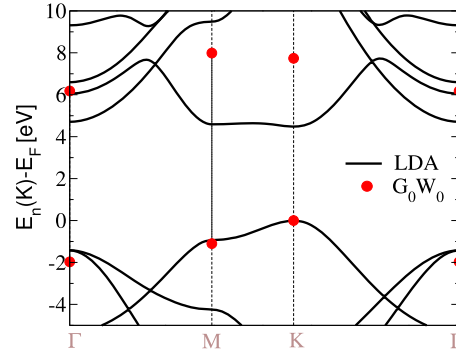


Figure 3. The band structure of hBN-ML obtained by the LDA. The red dots represent the q2D- G_0W_0 correction of highest occupied (valence) and the lowest unoccupied (conduction) bands in Γ , K and M points of Brillouin zone.

only 20 meV. However, it should be noted that the number of bands and CLFE energy cut-off are not independent [20, 24]. Thus, even if the number of bands is well converged, the low CLFE cut-off may be responsible for the disagreement. Figure 3 shows the hBN-ML band structure obtained using LDA. The red dots represent the q2D- G_0W_0 correction of highest occupied (or valence) $B(2p_z)$ and the lowest unoccupied (or conduction) $N(2p_z)$ bands in Γ , K and M points of Brillouin zone. It is worth noticing that after G_0W_0 correction the hBN possess the indirect band gap of 6.2 eV which slightly underestimate the values 6.58 eV reported in reference [24]. However, the q2D- G_0W_0 correction causes a mostly equal shift upwards of the conduction band which is also in accordance with previous, more detailed calculations [23, 24, 40, 41]. This allows us to use the scissor operator approach, i.e. at K point, the parabolic conduction band will be shifted only relative to parabolic valence band for the value of q2D- G_0W_0 band-gap. The LDA effective masses of the conduction and the valence bands at K point are $m_c^* = 0.8m_e$ and $m_v^* = 0.75m_e$, respectively.

2.2. Optical properties of q2D

One of the ways to describe the optical properties of a semiconducting system is to solve the 3D Bethe–Salpeter equation (BSE) for the electron–hole propagator. In the leading order of Coulomb interaction the BSE consists of an RPA term (creation and annihilation of electron–hole pairs), a ladder term (interaction between excited electron and hole) and of quasiparticle G_0W_0 corrections (renormalization of the electron and hole self-energies).

In the very first moment of the optical absorption process, the electron and the hole are created and they propagate for a short time without any scattering. Considering that the photon wavelength is usually much larger than q2D crystal thickness ($\lambda = 1/|\mathbf{Q}| \gg L$), such pure creation or annihilation processes can be described using an intrinsically 2D model. However, in a very short time after their creation, the electron and the hole being to interact with each other via the screened Coulomb interaction w and scatter with other excitations and impurities in the crystal. The short range and the local character of these scattering processes (they occur for larger wave vectors $|\mathbf{Q}'| \approx 1/a$) requires the microscopic treatment of the screened Coulomb interaction, i.e. its spatial dispersivity in all three dimensions becomes important. Because of that, the DFT-RPA screened Coulomb interaction w , implemented in the G_0W_0 self-energy and in the ladder term, should be calculated at a high level of accuracy, including a large number of unoccupied bands, as well as the 3D crystal-local field effects. For this reason, the 2D model is adequate at the RPA stage of calculation but inadequate when G_0W_0 and ladder corrections are included. However, below we still use the 2D approach because the short-range self-energy corrections are already implemented in band structure through the scissor operator approach. Moreover, considering the weak dependence of the ladder term on the number of bands, it is sufficient to include in the calculation only the scattering within one valence (v) and one conduction (c) band. Additionally, in the ladder term we restrict the calculation to only the strongest contribution $\mathbf{Q}' \approx 0$ which finally reduces to a 2D model.

2.2.1. Equation of motion for the electron–hole propagator

The dynamics of charge density fluctuation at wave vector \mathbf{Q} can be described by a Heisenberg equation for the charge density operator

$$\hat{\rho}(\mathbf{Q}) = \sum_{n,m} \sum_{\mathbf{K} \in \text{S.B.Z.}} \rho_{n\mathbf{K},m\mathbf{K}+\mathbf{Q}} \hat{c}_{n\mathbf{K}}^\dagger \hat{c}_{m\mathbf{K}+\mathbf{Q}} \quad (2.14)$$

where the charge vertices (2.4) are now $\rho_{n\mathbf{K},m\mathbf{K}+\mathbf{Q}} = \rho_{n\mathbf{K},m\mathbf{K}+\mathbf{Q}}(0)$. The Heisenberg equation for electron–hole excitations operator $\hat{\rho}_{\mathbf{K},\mathbf{K}+\mathbf{Q}}^{vc} = \hat{c}_{v\mathbf{K}}^\dagger \hat{c}_{c\mathbf{K}+\mathbf{Q}}$ is

$$i\hbar \frac{\partial}{\partial t} \hat{\rho}_{\mathbf{K},\mathbf{K}+\mathbf{Q}}^{vc} = [\hat{\rho}_{\mathbf{K},\mathbf{K}+\mathbf{Q}}^{vc}, H], \quad (2.15)$$

where we omit writing the spin index. The Hamiltonian of the system is given by

$$H = H_0 + H_{e-e}, \quad (2.16)$$

where the bare Hamiltonian is

$$H_0 = \sum_{n=v,c} \sum_{\mathbf{K} \in S.B.Z.} E_{\mathbf{K}}^n \hat{c}_{n\mathbf{K}}^\dagger \hat{c}_{n\mathbf{K}} \quad (2.17)$$

and electron–electron interaction is given by

$$H_{e-e} = -\frac{1}{2} \sum_{\mathbf{Q}} v_{\mathbf{Q}} \hat{\rho}^\dagger(\mathbf{Q}) \hat{\rho}(\mathbf{Q}). \quad (2.18)$$

Using the Wick theorem [42], the solution of the equation (2.15), to the first order in the Coulomb interaction $v_{\mathbf{Q}}$, can be written as

$$\begin{aligned} (\hbar\omega + E_{\mathbf{K}}^v - E_{\mathbf{K}+\mathbf{Q}}^c) \hat{\rho}_{\mathbf{K},\mathbf{K}+\mathbf{Q}}^{vc} &= \frac{2}{V} \sum_{\mathbf{K}'} v_{\mathbf{Q}} \rho_{c\mathbf{K}+\mathbf{Q},v\mathbf{K}} \rho_{v\mathbf{K}',c\mathbf{K}'+\mathbf{Q}} (f_{\mathbf{K}}^v - f_{\mathbf{K}+\mathbf{Q}}^c) \hat{\rho}_{\mathbf{K}',\mathbf{K}'+\mathbf{Q}}^{vc} \\ &+ \frac{1}{V} \sum_{\mathbf{K}'} v_{\mathbf{K}'-\mathbf{K}} \rho_{c\mathbf{K}+\mathbf{Q},c\mathbf{K}'+\mathbf{Q}} \rho_{v\mathbf{K}',v\mathbf{K}} (f_{\mathbf{K}+\mathbf{Q}}^c - f_{\mathbf{K}}^v) \hat{\rho}_{\mathbf{K}',\mathbf{K}'+\mathbf{Q}}^{vc} + \mathcal{O}(v^2). \end{aligned} \quad (2.19)$$

Equation (2.19) is a self-consistent integral equation, which is usually solved numerically. Various approximations have to be performed in order to obtain a more appealing analytical solution.

Equation (2.19) consists of three main contributions. The left-hand side term contains the DFT energies $E_{\mathbf{K}}^n$ which are already improved by the q2D-G₀W₀ self-energy. The first and the second term on the right-hand side represent RPA and ladder contributions, respectively. In the $\mathbf{Q} \approx 0$ limit the interband charge vertex $\rho_{v\mathbf{K},c\mathbf{K}+\mathbf{Q}} \propto \mathbf{Q}$ while the intraband is $\rho_{c\mathbf{K},c\mathbf{K}+\mathbf{Q}} \approx 1$. Hence the RPA term goes linearly with \mathbf{Q} and it can be put to zero. Secondly, this is the reason why we can neglect the ‘off-resonance’ scattering processes $\hat{\rho}_{\mathbf{K}',\mathbf{K}'+\mathbf{Q}}^{cv}$ in the higher order contribution to the ladder part in equation (2.19). Such ‘off-resonance’ processes that requires interband $v \leftrightarrow c$ transitions are neglected here in accordance with the so called the Tamm–Dancoff approximation [43].

We can also restrict our consideration to transitions which are only in narrow area around the K point of the Brillouin zone. Approximating the valence and the conduction bands by parabolas with effective masses m_v^* and m_c^* respectively, the left-hand side in (2.19) becomes

$$E_{\mathbf{K}}^c - E_{\mathbf{K}}^v \approx \Delta + \hbar^2 \mathbf{K}^2 / 2\mu. \quad (2.20)$$

Here $\mu = m_c^* m_v^* / (m_c^* + m_v^*)$ is the reduced mass and the band gap is Δ . By applying these approximations and defining the exciton energies Ω relative to the bottom of the conduction band, i.e. $\Omega = \hbar\omega - \Delta$, equation (2.19) becomes

$$(\Omega - \hbar^2 \mathbf{K}^2 / 2\mu) \hat{\rho}_{\mathbf{K},\mathbf{K}}^{vc} \approx -\frac{1}{V} \sum_{\mathbf{Q}'} v_{\mathbf{Q}'} \rho_{c\mathbf{K},c\mathbf{K}+\mathbf{Q}'} \rho_{v\mathbf{K}+\mathbf{Q}',v\mathbf{K}} \hat{\rho}_{\mathbf{K}+\mathbf{Q}',\mathbf{K}+\mathbf{Q}'}^{vc}. \quad (2.21)$$

In the above expression we have put $\mathbf{Q}' = \mathbf{K}' - \mathbf{K}$. Main contributions to the sum in (2.21) come from $\mathbf{Q}' \approx 0$ due to the singular behavior of the $v_{\mathbf{Q}'} \sim 1/|\mathbf{Q}'|$. In this regime the charge vertices can be approximated as $\rho_{v\mathbf{K}+\mathbf{Q}',v\mathbf{K}} \approx 1$ and equation (2.21) finally gives

$$(\Omega - \hbar^2 \mathbf{K}^2 / 2\mu) \hat{\rho}_{\mathbf{K},\mathbf{K}}^{vc} \approx -\frac{1}{V} \sum_{\mathbf{Q}' \approx 0} v_{\mathbf{Q}'} \hat{\rho}_{\mathbf{K}+\mathbf{Q}',\mathbf{K}+\mathbf{Q}'}^{vc}. \quad (2.22)$$

The particular set of higher order contributions to the Coulomb interaction in (2.22) can be provided by replacing the bare Coulomb interaction by a dynamically screened Coulomb interaction

$$v_{\mathbf{Q}'} \rightarrow w_{\mathbf{Q}'}(\omega).$$

Subsequently, the equation (2.22) is solved in the cases of bare and screened Coulomb potential.

2.2.2. 2D screened Coulomb interaction

In long wavelength limit $\mathbf{Q} \rightarrow 0$ the hBN-ML can be approximated as a fully 2D system and its independent electrons response matrix becomes

$$\lim_{\mathbf{Q} \rightarrow 0} \chi_{\mathbf{G}\mathbf{G}'}^0(\mathbf{Q}, \omega) = \chi^0(\mathbf{Q}, \omega) \delta_{\mathbf{G},0} \delta_{\mathbf{G}',0} \quad (2.23)$$

where

$$\chi^0(\mathbf{Q}, \omega) = L \chi_{\mathbf{G}=0, \mathbf{G}'=0}^0(\mathbf{Q}, \omega). \quad (2.24)$$

In this case the Dyson matrix equation (2.6) becomes a scalar equation with the solution $\chi = \chi^0 / (1 - v_{\mathbf{Q}} \chi^0)$ and the screened 2D Coulomb interaction becomes

$$w_{\mathbf{Q}}(\omega) = v_{\mathbf{Q}} / \varepsilon(\mathbf{Q}, \omega), \quad (2.25)$$

where the 2D dielectric function is

$$\varepsilon(\mathbf{Q}, \omega) = 1 - v_{\mathbf{Q}} \chi^0(\mathbf{Q}, \omega), \quad (2.26)$$

and where $v_{\mathbf{Q}} = 2\pi/|\mathbf{Q}|$. It is shown that in the long wavelength limit ($\mathbf{Q} \approx 0$) the hBN-ML 2D dielectric function can be approximated as [44]

$$\varepsilon(\mathbf{Q} \approx 0, \omega) = 1 + \lambda(\omega)|\mathbf{Q}|. \quad (2.27)$$

Inserting (2.27) into (2.25) the 2D dynamically screened Coulomb potential becomes

$$w_{\mathbf{Q}}(\omega) = \frac{2\pi}{|\mathbf{Q}|(1 + \lambda(\omega)|\mathbf{Q}|)}, \quad (2.28)$$

which after a Fourier transformation to the direct space becomes the Keldysh potential [49–51]

$$W(r) = \frac{e^2 \pi}{2\lambda} [Y_0(r/\lambda) - N_0(r/\lambda)]. \quad (2.29)$$

Here $Y_0(x)$ and $N_0(x)$ represent Struve and Neumann functions, respectively. Considering the wide hBN-ML band gap, the static approximation

$$\lambda(\omega) \approx \lambda(\omega = 0)$$

is also valid in the dynamic limit. The $\lambda(\omega = 0)$ is derived such that we first calculate the static ($\omega \approx 0$) 2D dielectric function (2.26) for different wave vectors Q . In the interval $Q < 0.2$ a.u. it behaves exactly as (2.27) such that the screening length $\lambda(\omega = 0)$ is derived by fitting (2.26) with (2.27) and extracting the slope parameter. Due to large hBN band-gap, the dynamical 2D dielectric function can be approximated as (2.27) for $\lambda(\omega = 0)$ up to $\hbar\omega \approx 4$ eV. This is the assumption we use in a further estimation of the optical properties of hBN. From (2.27), it is obvious that $\lambda(\omega = 0) = \lambda$ can also be interpreted as a screening length. The DFT screening length of hBN-ML is $\lambda = 12.6a_0$ [44]. However, given that λ strongly depends on the band-gap energy (λ decreases as the band-gap increases), in further calculations we use the value obtained using q2D-G₀W₀ corrected band-gap, which is $\lambda = 10a_0$.

2.2.3. Wannier model

Here we briefly mention the Wannier procedure for solving the equation (2.22) for the bare Coulomb interaction, in order to determine the exciton binding energy. After a Fourier transform of the electron–hole operator in the real space

$$\psi(\mathbf{r}) = \frac{1}{V} \sum_{\mathbf{K}} \hat{\rho}_{\mathbf{K},\mathbf{K}}^{vc} e^{i\mathbf{K}\cdot\mathbf{r}}, \quad (2.30)$$

(which, by assumption, is a smooth and well-behaved function of the electron–hole distance $|\mathbf{r}|$) it can be shown that equation (2.22) can be written as a two-body Schrödinger-like equation

$$\left[-\frac{\hbar^2}{2\mu} \nabla^2 - \frac{e^2}{|\mathbf{r}|} \right] \psi(\mathbf{r}) = \Omega \psi(\mathbf{r}). \quad (2.31)$$

The eigenvalues are well-known 2D hydrogen-like energies [52]

$$\Omega_n = -\frac{\mu}{m_e} \frac{1\text{Ry}}{(n-1/2)^2}, \quad (2.32)$$

where each state denoted by the quantum number n is $(n-1)$ times degenerated (not including the valley degeneracy which brings an extra factor of 2), since $\ell = 0, 1, \dots, n-1$. By inserting the reduced mass $\mu = 0.35m_e$ in (2.32) the hBN-ML ground state exciton energy can be estimated as $\Omega_1 = -1.4$ Ry (-19 eV). Thus obtained binding energy is highly overestimated because in (2.31) a bare Coulomb interaction is used. The binding energies will be more realistic when the interaction is replaced by a screened Coulomb interaction (2.29).

3. Results and discussion

3.1. Exciton energies and spatial extent

In this section we present the exciton binding energies obtained by solving the Schrödinger equation (2.31) for the screened potential (2.29) where the screening length $\lambda = 10a_0$ is obtained from DFT-RPA calculations. The exciton binding energies given in table 1 are presented in terms of two quantum numbers (n_r, ℓ) . The radial quantum number n_r gives the number of nodes in the radial part of the wave function and ℓ is the orbital quantum number. The states denoted by the $\ell = 0, 1, 2, \dots$ are labeled s, p, d, \dots . This is an adequate choice since the absolute square of the angular part of the wave function behaves like $|\Phi(\varphi)|^2 \sim \cos^2(\ell\varphi)$, thus resembling to the 2D projections of the 3D atomic hydrogen orbitals. Moreover, the principal quantum number $n = 1 + n_r + \ell$ can be introduced. Then the energy states labeled $\Omega(n_r, \ell)$ can be equally labeled $\Omega_{n\ell}$. For example, a state $\Omega(n_r = 1, \ell = 1)$ is equivalent to the state Ω_{3p} , etc. All energy states having the same n are given in the same colour in table 1.

The replacement of the bare Coulomb interaction by screened one (even wide band-gap hBN screening is not very efficient) causes strong reduction of the ground state exciton binding energy, from 19 eV (obtained from equation (2.32)) to $\Omega_{1s} = -2.53$ eV, as can be seen in table 1.

This result are similar to results recently obtained by combined *ab initio* and tight-binding Wannier calculation [23] although the discrepancies can be easily traced to the different values of the effective masses and especially to larger screening length ($\lambda \approx 20a_0$) used in reference [23]. Figure 5 shows the exciton (excitation) energies in hBN-ML how they would appear in, for example, an optical absorption experiment. The excitation energies are calculated using formula $\hbar\omega = \Delta + \Omega(n_r, \ell)$ where first ten binding energies (or energy levels) are taken from the table 1. The energy of 1s exciton $\hbar\omega \approx 5.14$ eV agrees very well with energy of $\mathbf{q} \approx 0$ singlet exciton $\hbar\omega \approx 5.3$ eV obtained by solving full *ab initio* Bethe–Salpeter equation [17]. Also we obtain excellent agreement with optical absorption [53, 54] or photoluminescence (PL) [55] experimental result, which reported $\hbar\omega_{1s} \approx 6$ eV.

The states with higher ℓ have lower energy for the same n . This can be seen from figure 5 in the case of $n = 3$ series $\Omega_{3d} < \Omega_{3p} < \Omega_{3s}$. This energy ordering is experimentally observed in the two-photon absorption experiments on tungsten disulphide [56]. The mean exciton radius, defined as the average electron–hole separation in the state $\psi_{n\ell}$, is calculated as $\bar{r}_{n\ell} = \langle \psi_{n\ell} | r | \psi_{n\ell} \rangle$. In the ground state $\bar{r}_{1s} \approx 6 a_0$ which is approximately equal to the unit cell dimension. The mean exciton radius decreases with $\Omega_{n\ell}$. For example, $\bar{r}_{2s} \approx 22 a_0$ and $\bar{r}_{2p} \approx 15 a_0$, while for the highest calculated energy level $\bar{r}_{4s} \approx 75 a_0$. In the figure 4 the screened potential $W(r)$ is plotted as a function of the screening length λ . As λ increases, the logarithmic nature of the potential becomes more apparent for small electron–hole separations. This can be seen by taking the two opposite limits of the expression (2.29)

$$W(r \rightarrow \infty) \rightarrow -e^2/r \quad (3.1)$$

$$W(r \rightarrow 0) \rightarrow (e^2/\lambda) \ln(r e^\gamma/2\lambda), \quad (3.2)$$

where γ is Euler–Mascherion constant. The shape of the above potential suggests the reason why the states within the same shell n differ in energy. The radial part of the wave function of the states with lower ℓ is located closer to the origin where it is governed by a weaker logarithmic potential (3.2). Therefore, the binding energy is smaller and the mean radius is larger compared to the states with larger values of ℓ which are under the influence of stronger bare Coulomb potential (3.1) and thus have larger binding energy and smaller spatial extension.

Table 1. The first ten exciton energy levels in eV. The states within the same shell are given in the same colour: $n = 1, 2, 3, 4$ in maroon, red, orange and olive green, respectively.

$ \Omega(n_r, \ell) $	$\ell = 0$	$\ell = 1$	$\ell = 2$	$\ell = 3$
$n_r = 0$	2.53	1.09	0.57	0.34
$n_r = 1$	0.85	0.5	0.32	
$n_r = 2$	0.42	0.29		
$n_r = 3$	0.25			

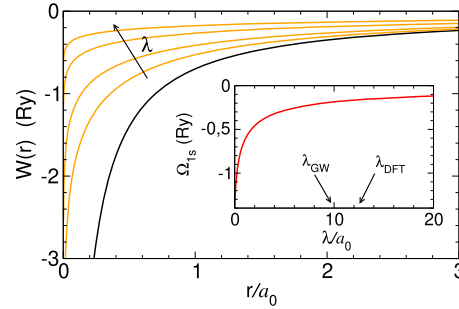


Figure 4. The bare Coulomb potential (black) and the screened Coulomb potential (2.29) (orange) plotted for different parameters $\lambda = 1, 2, 5, 10a_0$. The insert shows dependence of the ground state exciton binding energy Ω_{1s} on λ for $\mu = 0.35m_e$.

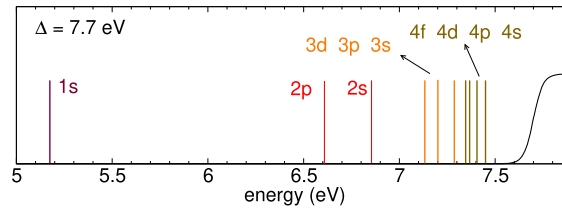


Figure 5. The excitonic spectrum of hBN-ML containing first ten levels (solid lines) from the table 1. The dashed maroon line represents the exciton ground state energy $\Omega_{1s}(\lambda_{\text{DFT}}) = -2.53$ eV calculated using the λ_{DFT} static screening length. The threshold energy for the single particle excitations is represented by a black line.

Here, it is also suggestive to see how the ground state exciton binding energy Ω_{1s} depends on the screening length λ . This dependence is shown in the insert of figure 4. In the limit $\lambda \rightarrow 0$ the $\Omega_{1s}(\lambda) = -19$ eV as obtained from equation (2.32). However, with the increase of the screening length, the exciton binding energy rapidly decreases such that for $\lambda_{\text{GW}} = 10a_0$ it is $\Omega_{1s}(\lambda) = -2.53$ eV, as already discussed in table 1. In the opposite limit ($\lambda \rightarrow \infty$) a saturation of the $\Omega_{1s}(\lambda)$ can be seen. The analytical approximation of $\Omega_{1s}(\lambda \rightarrow \infty)$ will be considered in the last section.

Observing figure 5, some conclusions can be made regarding the interaction of excitons with phonons, impurities, and electrons, changing their appearance in the absorption spectrum. Due to these interactions, the exciton signals will be broader and slightly shifted in energy. This causes an overlap between the individual exciton signals that are close in energy, to the point that they can even be joined with the single particle threshold. This would imply the indistinguishability of partial contributions originating from the single particle excitations and the excitons in the optical absorption spectra.

3.2. The single particle gap estimation

The results obtained in section 3.1 can be used to estimate the hBN-ML band gap which is experimentally still not observed and *ab initio* results are not yet in consensus, considering that the values between 7 and 9 eV are reported [14, 15]. The band gap Δ' can be estimated by finding a value Δ' for which the theoretically obtained exciton energy $\hbar\omega_{1s}$ is equal to the experimental value $\hbar\omega_{1s}^{\text{exp}}$ or using the formula

$$\hbar\omega_{1s}^{\text{exp}} = \Delta' + \Omega_{1s}[\lambda(\Delta')]. \quad (3.3)$$

This equation can be solved numerically by calculating the binding energy Ω_{1s} as a function of Δ' which is then inserted in (3.3). However, here we take into account that the binding energy Ω_{1s} changes slowly for screening lengths in the range of $\lambda = 10a_0 \pm 2a_0$ which essentially covers the values of the static screening

length calculated for the bandgap between 4.5 eV and 10 eV, as shown in figure 4 insert. Hence, for a reliable estimate we can choose the screening length value $\lambda = 10a_0$ obtained from the G_0W_0 band structure. Moreover, the same value for exciton binding energy $\Omega_{1s}(\lambda) = -2.53$ eV can be used. In optical absorption or PL experiments on quasi hBN-ML [53–55] the exciton ground state signal appears at the energy $\hbar\omega_{1s}^{\text{exp}} \approx 6$ eV. Therefore using this data and (3.3) the quasiparticle band gap is estimated as $\Delta' \approx 8.5$ eV, which overestimate the q2D- G_0W_0 band gap $\Delta_{\text{GW}} = 7.7$ eV, as presented in section 2.1.

3.3. The limit of high polarizability—EBK procedure

Here we present the analytical form for the ground state exciton energy Ω_{1s} in the limit of high screening length ($\lambda \rightarrow \infty$). As the *ab initio* results predict, hBN-ML can not be considered a material where the above condition applies, since $\lambda = 10a_0$. However, other two-dimensional materials, such as transition-metal dichalcogenides, have a much larger λ . We have also performed the DFT calculations for molybdenum disulfide (MoS₂-ML) and molybdenum ditelluride (MoTe₂-ML) and obtained $\lambda = 76a_0$ and $120a_0$ and $\mu = 0.25m_e$ and $0.3m_e$, respectively. The relatively large λ for these two materials originates from their spatial structure where a transition metal plane lies between the planes of two chalcogenide atoms. This geometrical coordination reduces the electronic hopping elements between the atomic orbitals of the neighbouring atoms [57], therefore causing the flattening of the electron bands. Smaller dispersivity of the electron bands combined with the smaller band gap, such as $\Delta_{\text{DFT}} \approx 1.8$ eV in the case of MoS₂-ML and $\Delta_{\text{DFT}} \approx 1.2$ eV in the case of MoTe₂-ML, give a static screening length almost an order of magnitude larger compared to $\lambda = 10a_0$ in hBN-ML. This will certainly be responsible for the logarithmic behaviour of the screened potential over a sizeable electron–hole spatial extension, as can be seen in figure 4, making the logarithmic potential (3.2) adequate for determining the ground state exciton binding energy.

In this case the semiclassical EBK approach can be applied to calculate low-lying ($\ell = 0$) exciton energy levels. The general EBK approximation [58] asserts that in the spherical symmetric problem the phase integral of the radial impulse is quantized as

$$2 \int_0^{r_0} \sqrt{2\mu [\Omega - W(r \rightarrow 0)]} dr = \pi \hbar (n_r + z/4). \quad (3.4)$$

Here r_0 is the classical turning point, $n_r = 0$ is the radial quantum number for the ground state energy case, and $z = 2$ is the Maslov index [58], which gives the number of classical turning points. This leads to the implicit expression for the exciton ground state energy

$$\frac{\sqrt{\pi}}{2} \text{Erf} \left(\sqrt{\Omega \lambda / e^2} \right) e^{\Omega \lambda / e^2} - \sqrt{\Omega \lambda / e^2} = \frac{\pi \hbar e^\gamma z / 4}{\sqrt{32 \mu e^2 \lambda}}. \quad (3.5)$$

In the case of a large screening length ($\lambda / e^2 \rightarrow \infty$), when the exponential function dominates and the error function is $\text{Erf}(x \rightarrow \infty) = 1$, equation (3.5) can be simplified, which finally leads to the analytical expression for the ground state exciton binding energy

$$\Omega_{1s}(\lambda) \approx \frac{e^2}{\lambda} \ln \left(\frac{\hbar e^\gamma z \sqrt{\pi}}{8 \sqrt{2 \lambda \mu e^2}} \right). \quad (3.6)$$

The similar expression has been found as the limiting solution of the Schrödinger equation for the logarithmic potential [59], while the results of reference [60] predict $\Omega_{1s}(\lambda) \approx -(3/4\pi)e^2/\lambda$ behaviour in the same limit suggesting that they neglected the logarithmic term. The expression (3.6) gives a fairly accurate description of $\Omega_{1s}(\lambda)$ in $\lambda \rightarrow \infty$ limit, but it can be improved by setting $z = 3.5$. For that purpose we introduce the dimensionless values λ' and μ' through $\lambda = \lambda' a_0$ and $\mu = \mu' m_e$ and by putting $z = 3.5$ in the expression (3.6) we have

$$\Omega_{1s}(\lambda) \approx -\frac{1\text{Ry}}{\lambda'} \ln(\lambda' \mu'). \quad (3.7)$$

Figure 6 shows the comparison between the ground state energy $\Omega_{1s}(\lambda)$ obtained using the static screened potential (2.29) (solid lines) and using the analytical form (3.7) (dashed lines), for two different reduced masses μ . It is evident that the agreement between the (solid and dashed) curves starts only for large values of λ .

Using formula (3.3) outlined in section 3.2, together with formula (3.6) and the experimentally determined exciton energies $\hbar\omega_{1s}^{\text{exp}}(\text{MoS}_2) \approx 1.9$ eV and $\hbar\omega_{1s}^{\text{exp}}(\text{MoTe}_2) \approx 1.2$ eV [61–63], it is possible to estimate the single particle band gap of MoS₂ and MoTe₂ monolayers. Inserting the calculated polarizabilities and the reduced masses into (3.6) and using (3.3) gives $\Delta'_{\text{MoS}_2} \approx 2.5$ eV and

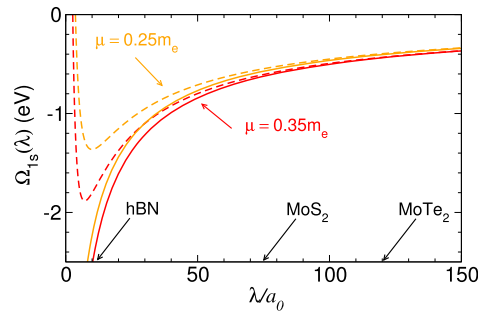


Figure 6. Ground state exciton binding energy Ω_{1s} as a function of the screening length λ obtained by solving the Schrödinger equation for the statically screened potential (2.29) (solid lines) and using the analytical expression equation (3.6) (dashed lines) with $z = 3.5$ for two values of the reduced masses $\mu = 0.35m_e$ (red) and $\mu = 0.25m_e$ (orange). The *ab initio* values of the static screening lengths of hBN, MoS₂ and MoTe₂ monolayers are indicated.

Table 2. The quasiparticle band gap Δ' estimated from screening lengths λ , reduced masses μ , the experimental exciton energies $\hbar\omega_{1s}^{\text{exp}}$ [61–63] and equations (3.3) and (3.7). The Δ represents the band gaps obtained using q2D-G₀W₀ method (2.1)–(2.12).

2D crystal	hBN	MoS ₂	MoTe ₂
λ/a_0	10	76	120
μ/m_e	0.35	0.25	0.3
$\hbar\omega_{1s}^{\text{exp}}$	6.0	1.9	1.2
Ω_{1s}	−2.53	−0.53	−0.4
Δ'	8.5	2.43	1.6
Δ_{GW}	7.7(7.37 ^a)	2.6(2.78 ^b)	1.9(1.77 ^c)

^aThe values in parentheses are taken from reference [24].

^bThe values in parentheses are taken from reference [20]

^cThe values in parentheses are taken from reference [25].

$\Delta'_{\text{MoTe}_2} \approx 1.6$ eV. The DFT polarizabilities, the experimental exciton energies and the estimated quasiparticle band-gap for the three studied 2D crystals are summarized in table 2. It is worth noticing that the estimated band-gaps Δ' agree satisfactorily well with other BSE-GW calculations [18–20, 25–27] considering that the relative error is not larger than 15%, while the method used to derive Δ' is very simple. The q2D-G₀W₀ band-gaps Δ_{GW} in hBN, MoS₂ and MoTe₂ are also compared with other G₀W₀ calculations. The hBN direct gap ($K \rightarrow K$) 7.7 eV slightly overestimates and the indirect gap ($K \rightarrow \Gamma$) 6.2 eV (shown in figure 3) slightly underestimates the values 7.37 eV and 6.58 eV, respectively, reported in reference [24]. The MoS₂ band gap of 2.6 eV slightly underestimates the value 2.78 eV reported in reference [20]. As in reference [20], there is a problem here concerning the band gap in the M point which is 4.8 eV instead of 3.9 eV obtained by using 10^4 bands and a 500 eV CLFE cut-off. However, the high-resolution STM spectroscopy measurements [64, 65] show that MoS₂ monolayer band-gap is in the range 2.2–2.4 eV. This substantial reduction is probably a result of MoS₂ monolayer being deposited on graphite which due to metallic screening reduces the band gap. Finally the MoTe₂ band-gap 1.9 eV overestimates the other results which are in range 1.72–1.77 eV [25–27]. The disagreement is probably because here the CLFE are included only in the perpendicular (z) direction which simultaneously results in the band gap converging with fewer bands.

4. Conclusion

The alternative q2D-G₀W₀-BSE method was applied to calculate the quasiparticle band-gap and the exciton binding energies in hBN, MoS₂ and MoTe₂. The q2D-G₀W₀-BSE method is complementary to existing high performance GW-BSE methodology because it is computationally less demanding (at the G₀W₀ stage the CLFE are included only in the direction of broken periodic symmetry), it excludes the spurious Coulomb interaction and takes advantage of the reduction of BSE to its 2D counterpart. The obtained results for hBN, MoS₂ and MoTe₂ single-particle band-gap and exciton binding energies agree well with other theoretical results. Using the calculated ground state exciton binding energies Ω_{1s} and the experimental exciton (excitation) energies $\hbar\omega_{1s}$ the realistic value of the single particle band gap Δ' in hBN, MoS₂ and MoTe₂ monolayers are estimated and compared with other results. An analytical expression for the ground

state exciton binding energy Ω_{1s} is obtained in the high polarizability limit using the EBK procedure. This was shown to be valid for the family of TMDs and therefore can be a very useful tool for experimentalists.

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Appendix A. Transformation from 3D to q2D dynamical response

The RPA time-ordered irreducible polarizability defined as (2.7) is periodic in the direction perpendicular to q2D crystal plane (the z direction), i.e. it periodically repeats from supercell to supercell, separated by the distance L . Therefore, the time-ordered response function calculated by solving the RPA Dyson equation

$$\chi_{\mathbf{G}\mathbf{G}'}^T(\mathbf{q}, \omega) = \tilde{\chi}_{\mathbf{G}\mathbf{G}'}^0(\mathbf{q}, \omega) + \sum_{\mathbf{G}_1\mathbf{G}_2} \tilde{\chi}_{\mathbf{G}\mathbf{G}_1}^0(\mathbf{q}, \omega) v_{\mathbf{G}_1\mathbf{G}_2}^{3D}(\mathbf{q}) \chi_{\mathbf{G}_2\mathbf{G}'}^T(\mathbf{q}, \omega), \quad (\text{A.1})$$

where

$$v_{\mathbf{G}\mathbf{G}'}^{3D}(\mathbf{Q}) = \frac{4\pi}{|\mathbf{Q} + \mathbf{G}|^2} \delta_{\mathbf{G}\mathbf{G}'}, \quad (\text{A.2})$$

contains the effects of interaction with surrounding q2D crystals. However, the interaction with the surrounding supercells can be easily completely eliminated. Instead of performing the complete Fourier transform (A.1) and (A.2), we can start from the partially Fourier-transformed RPA Dyson equation

$$\begin{aligned} \chi_{\mathbf{G}_{\parallel}\mathbf{G}'_{\parallel}}^T(\mathbf{Q}, \omega, z, z') &= \tilde{\chi}_{\mathbf{G}_{\parallel}\mathbf{G}'_{\parallel}}^0(\mathbf{Q}, \omega, z, z') + \sum_{\mathbf{G}_{\parallel 1}} \int_{-L/2}^{L/2} dz_1 dz_2 \\ &\times \tilde{\chi}_{\mathbf{G}_{\parallel}\mathbf{G}_{\parallel 1}}^0(\mathbf{Q}, \omega, z, z_1) v_{\mathbf{G}_{\parallel 1}}^{2D}(\mathbf{Q}, z_1, z_2) \chi_{\mathbf{G}_{\parallel 1}\mathbf{G}'_{\parallel}}^T(\mathbf{Q}, \omega, z_2, z'). \end{aligned} \quad (\text{A.3})$$

Since the z coordinate remains untransformed, the 2D Fourier transform of the bare Coulomb interaction becomes

$$v_{\mathbf{G}_{\parallel}}^{2D}(\mathbf{Q}, z, z') = \frac{2\pi}{|\mathbf{Q} + \mathbf{G}_{\parallel}|} e^{-|\mathbf{Q} + \mathbf{G}_{\parallel}||z - z'|} \quad (\text{A.4})$$

and

$$\tilde{\chi}_{\mathbf{G}_{\parallel}\mathbf{G}'_{\parallel}}^0(\mathbf{Q}, \omega, z, z') = \frac{1}{L} \sum_{G_z G'_z} \tilde{\chi}_{\mathbf{G}\mathbf{G}'}^0(\mathbf{Q}, \omega) e^{iG_z z - iG'_z z'} \quad (\text{A.5})$$

represents the Fourier expansion of the irreducible polarizability in z and z' direction. Since the z_1 and z_2 integrations in (A.3) are performed from $-L/2$ to $L/2$, instead of from $-\infty$ to ∞ , as done in (A.1), the interaction between density fluctuations is possible only within the same layer, while the interaction with the polarization in surrounding layers is completely excluded. We can say that we ‘closed’ the Coulomb interaction within unit cell $-L/2 < z < L/2$. After inserting the Fourier expansion (A.5), and a similar one for χ^T , in the RPA Dyson equation (A.3) we can perform integration over z_1 and z_2 and it again becomes a full matrix equation

$$\chi_{\mathbf{G}\mathbf{G}'}^T(\mathbf{Q}, \omega) = \tilde{\chi}_{\mathbf{G}\mathbf{G}'}^0(\mathbf{Q}, \omega) + \sum_{\mathbf{G}_1\mathbf{G}_2} \tilde{\chi}_{\mathbf{G}\mathbf{G}_1}^0(\mathbf{Q}, \omega) v_{\mathbf{G}_1\mathbf{G}_2}^{2D}(\mathbf{Q}) \chi_{\mathbf{G}_2\mathbf{G}'}^T(\mathbf{Q}, \omega), \quad (\text{A.6})$$

similar to Dyson equation (A.1). However, now matrix of the bare Coulomb interaction becomes

$$v_{\mathbf{G}\mathbf{G}'}^{2D}(\mathbf{Q}) = v_{\mathbf{G}\mathbf{G}'}^{3D}(\mathbf{Q}) - p_{G_z} p_{G'_z} \frac{4\pi(1 - e^{-|\mathbf{Q} + \mathbf{G}_{\parallel}|L})}{|\mathbf{Q} + \mathbf{G}_{\parallel}|L} \frac{|\mathbf{Q} + \mathbf{G}_{\parallel}|^2 - G_z G'_z}{(|\mathbf{Q} + \mathbf{G}_{\parallel}|^2 + G_z^2)(|\mathbf{Q} + \mathbf{G}_{\parallel}|^2 + G_z'^2)} \delta_{\mathbf{G}_{\parallel}\mathbf{G}'_{\parallel}}, \quad (\text{A.7})$$

where

$$p_{G_z} = \begin{cases} 1; & G_z = \frac{2k\pi}{L} \\ -1; & G_z = \frac{(2k+1)\pi}{L}, \quad k = 0, 1, 2, 3, \dots \end{cases}$$

This means that the time-ordered response function in q2D crystals can be obtained the same way as in the 3D (bulk) calculation (A.1), with the matrix of the bare Coulomb interaction (A.2) replaced by (A.7), i.e.

$$v_{\mathbf{G}\mathbf{G}'}^{3\text{D}}(\mathbf{Q}) \rightarrow v_{\mathbf{G}\mathbf{G}'}^{2\text{D}}(\mathbf{Q}).$$

ORCID iDs

Vito Despoja  <https://orcid.org/0000-0002-6159-9876>

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