Optical schemes for quantum computation in quantum dot molecules

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We give three methods for entangling quantum states in quantum dots. We do this by showing how to tailor the resonant energy (Förster-Dexter) transfer mechanisms and the biexciton binding energy in a quantum dot molecule. We calculate the magnitude of these two electrostatic interactions as a function of dot size, interdot separation, material composition, confinement potential, and applied electric field by using an envelope function approximation in a two-cuboid dot molecule. In the first implementation, we show that it is desirable to suppress the Förster coupling and to create entanglement by using the biexciton energy alone. We show how to perform universal quantum logic in a second implementation which uses the biexciton energy together with appropriately tuned laser pulses: by selecting appropriate material parameters high-fidelity logic can be achieved. The third implementation proposes generating quantum entanglement by switching the Förster interaction itself. We show that the energy transfer can be fast enough in certain dot structures that switching can occur on a time scale which is much less than the typical decoherence times.

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I. INTRODUCTION

Quantum dots^{1,2} are quantum heterostructures which are composed of nanoscale regions of one type of material which is embedded in a second type. In a semiconductor quantum dot (QD), materials with differing band gaps are used; this leads to the possibility of electronic confinement within the dot region. Moreover, the confined electronic states can be accurately controlled by varying the dot size, shape or composition, and the number of confined electrons; all of these may be altered by using different growth conditions and hence specifically tailored "artificial atoms" or "superatoms" can be produced.^{1,2} Some prominent atomlike properties of QD's include an electronic shell structure,³ Rabi oscillations,⁴ photon antibunching,^{5,6} controlled quan-tum light emission,^{7,8} and quantum entanglement.^{9,10} One of the most intriguing possible applications of quantum dots is that they may be used to build quantum computers.^{11,12} A practical realization of a quantum computer would be very significant, since there exist theoretical quantum algorithms which would make some classically hard computational problems tractable.¹³ Such quantum devices could also accurately simulate any physical system (and the evolution of its local interactions) by invoking the same amount of energy and Hilbert space requirements as the system itself.^{14–16}

The basic unit of a quantum computer is a two-level quantum system, the so-called qubit. Of the utmost importance is the identification of a physical system where a coherent qubit evolution can be performed, thus allowing a precise execution of the elementary quantum gates required for universal quantum computation.^{11,12} Many different types of hardware for embodying qubits have been proposed (for a collection of papers detailing some of these see Ref. 17) and some of them have already been implemented for performing elementary quantum gate operations. These include ion traps,^{18–22} quantum electrodynamics

cavities,^{23–27} nuclei in molecules,^{28–31} dopants in semiconductors,^{32–34} optical lattices and Bose-Einstein condensates,^{35–37} Josephson junctions,^{38–41} and quantum dots.^{10,42-51} In this paper we concentrate on a quantum dot implementation. Previous proposals^{9,10,42–44,46,47,51–53} include the use of a single electron 43,51,52 or nuclear 44 spin located on each of an array of interacting dots, the presence or absence of an electron charge state,⁴² or the use of excitonic states.^{10,46,47,53} We show how an energy selective approach to manipulating the excitonic states of coupled QD's, together with control over the energy transfer and biexciton binding energy, can be used to perform quantum computation (QC) and to produce controlled exciton quantum entanglement. In so doing, we investigate the Förster-Dexter resonant energy transfer, a mechanism first studied in the context of the sensitized luminescence of solids, 54,55 in which an excited sensitizer atom can transfer its excitation to a neighboring acceptor atom, via an intermediate virtual photon. This mechanism is also responsible for photosynthetic energy processes in antenna complexes, biosystems (BS's) that harvest sunlight.⁵⁶ More recently, interest has focussed on energy transfer in quantum dot nanostructures⁵⁷ and within molecular systems (MS's).⁵⁸ In this paper we show how to exploit such energy-transfer mechanisms with a view to processing quantum information. This paper is a more detailed account of the work which appears in Ref. 46.

II. BUILDING QUANTUM LOGIC GATES

We consider the Hamiltonian of two interacting quantum dots. We assume that the dots are sufficiently far apart that tunneling processes between them may be neglected but that there is a strong exciton-exciton coupling. Our two-level system is represented in each dot by a single low-lying exciton state $|1\rangle$ and the ground state $|0\rangle$. Then the interaction

Hamiltonian can be written in the computational basis $(\{|00\rangle, |01\rangle, |10\rangle, |11\rangle\}$, with the first digit referring to dot I and the second to dot II) as follows:

$$\hat{H} = \begin{pmatrix} \omega_0 & 0 & 0 & 0 \\ 0 & \omega_0 + \omega_2 & V_F & 0 \\ 0 & V_F & \omega_0 + \omega_1 & 0 \\ 0 & 0 & 0 & \omega_0 + \omega_1 + \omega_2 + V_{XX} \end{pmatrix}.$$
(1)

The diagonal interaction V_{XX} is the direct Coulomb binding energy between two excitons, one located on each dot, and $V_{\rm F}$ denotes the Coulomb exchange (Förster) interaction which is off-diagonal and therefore induces the transfer of an exciton from one QD to the other. These are the only Coulomb interaction terms which act between the qubits and will be calculated and discussed in detail in Sec. V. ω_0 denotes the ground-state energy, ω_1 (ω_2) refers to the energy required to create an exciton on dot I (II) in the absence of interactions, and includes intradot coupling contributions (direct Coulomb binding energy and spin splitting) which we shall discuss in Sec. IV. We also define $\Delta_0 \equiv \omega_1 - \omega_2$ to be the difference between the exciton creation energy for dot I and that for dot II in the absence of interactions between the dots. Thus, if $H_0 = H_I + H_{II}$ denotes the free particle Hamiltonian, then $H_0(|\gamma_1\rangle|\gamma_2\rangle) = (\gamma_1\omega_1 + \gamma_2\omega_2)|\gamma_1\rangle|\gamma_2\rangle, \ \gamma_1, \gamma_2$ =0,1 (\hbar =1 throughout this paper), then $H=H_0+V_{1,2}$ (where $V_{1,2}$ accounts for the qubit-qubit interactions, V_{XX} and $V_{\rm F}$) is the system's overall Hamiltonian. In the case of an n-qubit register with nearest-neighbor interactions, the Hamiltonian takes the form $H = H_0^{(n)} + \sum_{i=1}^{n-1} V_{i,i+1}$, where $H_0^{(n)} \equiv \sum_{i=1}^n H_i$ is the free particle Hamiltonian, and $V_{i,i+1}$ are the interaction terms. A related Hamiltonian was investigated in Ref. 53, but there the off-diagonal interaction terms $(V_{\rm F})$ were neglected.

The eigenenergies and eigenstates of the interacting qubit system are

$$E_{00} = \omega_{0}, \quad |\Psi_{00}\rangle = |00\rangle,$$

$$E_{01} = \omega_{0} + \omega_{1} - \frac{\Delta_{0}}{2}(1+A), \quad |\Psi_{01}\rangle = c_{1}|10\rangle + c_{2}|01\rangle,$$

$$E_{10} = \omega_{0} + \omega_{1} - \frac{\Delta_{0}}{2}(1-A), \quad |\Psi_{10}\rangle = -c_{1}|01\rangle + c_{2}|10\rangle,$$

$$E_{11} = \omega_{0} + \omega_{1} + \omega_{2} + V_{\text{YX}}, \quad |\Psi_{11}\rangle = |11\rangle, \quad (2)$$

where $A = \sqrt{1 + 4(V_F/\Delta_0)^2}$, $c_1 = \sqrt{(A-1)/2A}$ ($\approx V_F/\Delta_0$ for $V_F/\Delta_0 \ll 1$) and $c_2 = \sqrt{(A+1)/2A}$. The eigenenergies in the absence and presence of interdot interactions are displayed in Figs. 1(a), 1(b), and 1(c). Figure 1(a) shows the energy levels when the interactions are off; Fig. 1(b) shows these when the interactions are on, but where $V_F \ll \Delta_0$; Fig. 1(c) shows E_{10} and E_{01} as a function of the ratio V_F/Δ_0 . Figure 1(d) shows c_1 and c_2 as a function of V_F/Δ_0 . These figures demonstrate that V_F causes a mixing of the states $|01\rangle$ and $|10\rangle$ such that the eigenstates of the interacting system are not the same as the computational basis. As we show below, this V_F coupling can be used for generating highly entangled states.

Single-qubit operations can be achieved by inducing Rabi oscillations in the excitonic system (e.g., see Refs. 4 and 59). If we take a Bloch sphere representation of a qubit, where the state $|0\rangle$ is represented by a unit vector from the origin to the north pole of the Bloch sphere and the state $|1\rangle$ by a unit vector to the south pole, then the qubit state $|\psi\rangle$ $= \exp(i\lambda) [\cos(\theta/2)|0\rangle + \exp(i\varphi)\sin(\theta/2)|1\rangle],$ where λ is a phase, is defined by the global unit vector $(\cos \varphi \sin \theta, \sin \varphi \sin \theta, \cos \theta)$. Depending an the values of θ and φ , this vector may point to any point on the surface of the sphere, and for universal quantum computation it must be possible to move the vector between any two of these points; this defines an arbitrary single-qubit rotation. In our QD system, this control can be achieved by using laser pulses to induce two distinct Rabi oscillations (see Fig. 2). The energy and length of such pulses must take into account structural factors such as the dot confinement energies and transition



FIG. 1. A schematic diagram of the properties of the model Hamiltonian, Eq. (1). (a) Energy levels in the absence of qubit (interdot) interactions. (b) Energy levels in the presence of qubit interactions for two dots (I and II) of different excitation frequencies. $\epsilon_{12} = \omega_2 + V_{XX} - \delta$, $\epsilon_{21} = \omega_1 + V_{XX} + \delta$, and $\delta \equiv V_F^2/\Delta_0$, where V_F and V_{XX} represent the strengths of the Förster and direct Coulomb binding interactions, respectively. In this case $V_F \ll \Delta_0$. (c) Eigenenergies E_{01} and E_{10} corresponding to the qubit eigenstates $|\Psi_{01}\rangle$ and $|\Psi_{10}\rangle$ as a function of the ratio V_F/Δ_0 for $\omega_1/\Delta_0 \equiv 20$. For comparison, the dashed lines show the energies when $V_F/\Delta_0 = 0$. (d) The eigenstate coefficients c_i as a function of the strength V_F/Δ_0 . These coefficients show the departure from the basis states $|01\rangle$ and $|10\rangle$ followed by the eigenstates $|\Psi_{01}\rangle = c_1|10\rangle + c_2|01\rangle$ and $|\Psi_{10}\rangle = -c_1|01\rangle + c_2|10\rangle$.



FIG. 2. Bloch sphere representation of a single qubit. In order to perform an arbitrary single-qubit rotation in the exciton system, it is necessary to have the ability to induce two different Rabi oscillations. Example trajectories for two such oscillations are shown by dashed lines.

dipole moments.^{59,60} It is also essential that the exciton states have long enough decoherence times that control over the phase φ is possible.^{61,62} For example, self-assembled semiconductor (e.g., InGaAs/GaAs) quantum dots could be advantageous for qubit manipulations since they exhibit large dipole moments and long dephasing times.^{4,59} We shall return to the role of the QD material composition parameters for QC below.

The $V_{\rm XX}$ and $V_{\rm F}$ interactions lead to three possible ways of achieving quantum entanglement. First, if the ratio $V_{\rm F}/\Delta_0 \gg 1$, the eigenstates of the system are approximately $|00\rangle$, $1/\sqrt{2}(|10\rangle - |01\rangle)$, $1/\sqrt{2}(|10\rangle + |01\rangle)$, and $|11\rangle$. We now further assume that the ratio $V_{\rm F}/\Delta_0$ can be controlled, by means of applying an electric field to either change $V_{\rm F}$ directly, or to increase Δ_0 by means of the Stark shift (we shall discuss both of these effects in detail in Sec. V A). Then, we initially prepare the system in a state where $V_{\rm F}/\Delta_0 \ll 1$ and we selectively excite QD I and create $|10\rangle$. Now, when the Förster interaction is turned on, the system will naturally evolve sequentially into the following states: $|10\rangle \mapsto 1/\sqrt{2}(|10\rangle + i|01\rangle) \mapsto |01\rangle \mapsto 1/\sqrt{2}(|10\rangle - i|01\rangle) \mapsto |10\rangle$ (see Fig. 3). This evolution could be then stopped when the system is in a maximally entangled state by applying an electric field to suppress the Förster coupling once more, an effect which we shall again explore in detail in Sec. V A.

Second, if the system does not have a strong Förster coupling, i.e., $V_F/\Delta_0 \ll 1$ all the time, the computational basis states are essentially the eigenstates of the system. Then the V_{XX} coupling term implies that the resonant frequency for transitions between the basis states $|0\rangle$ and $|1\rangle$ of one qubit *depends* on the state of the neighboring qubit. This means that it is possible to construct a controlled-NOT gate in this system. Such a gate flips the target qubit *k* if the control qubit *j* is in the state $|1\rangle$ and acts trivially otherwise: $CNOT_{jk}(|m\rangle_j|n\rangle_k) \mapsto |m\rangle_j |m \oplus n\rangle_k$, where $m, n \in \{0,1\}$, and \oplus denotes addition modulo 2 or XOR operation. By referring to Fig. 1(b), we can see that the logic operation $CNOT_{12}(|1\rangle_1|0\rangle_2) \mapsto |1\rangle_1|1\rangle_2$ can be achieved by illuminating the qubit system in the state $|10\rangle$ with a π pulse of



FIG. 3. Illustration of how to create an entangled state by manipulating the off-diagonal Förster coupling between two nanostructures. When the interaction is on and much greater than Δ_0 , the eigenstates of the system correspond to the large dots on the equator of the Bloch sphere. Thus after selective excitation of state $|10\rangle$ the system will naturally evolve to the equator. The interaction may then be suppressed by means of an applied field, and the state remains maximally entangled as the eigenstates become the computational basis states.

energy $\epsilon_{12} = \omega_2 + V_{XX} - \delta$ (a pulse which we label $\pi_{\epsilon_{12}}$). Conversely, if the role of the control qubit is to be performed the second qubit, the gate operation $(\text{NOT}_{21}(|0\rangle_1|1\rangle_2) \mapsto |1\rangle_1|1\rangle_2$ can be realized via the application of a π pulse of energy $\epsilon_{21} = \omega_1 + V_{XX} + \delta$ (a $\pi_{\epsilon_{21}}$ pulse) to the system state $|01\rangle$. Crucially, the energy difference between these two π pulses, and hence the energy selectivity of the logic gate, is determined by Δ_0 , V_{XX} , and V_F . From Fig. 1(b) we can also see how to use the controlled-NOT gate to create maximally entangled states. For example, if we start in the ground state and first apply a $\pi/2$ or $3\pi/2$ pulse at energy ω_1 , we create the states $1/\sqrt{2}(|00\rangle \pm |10\rangle)$; if we now apply a $\pi_{\epsilon_{12}}$ pulse, we generate the maximally entangled states $1/\sqrt{2}(|00\rangle \pm |11\rangle).$

Third, the interaction with the laser field does not necessarily have to be the entangling mechanism in the case where $V_{\rm F}/\Delta_0 \ll 1$. If each of two neighboring single qubits are prepared in the superposition state $1/\sqrt{2}(|0\rangle + |1\rangle)$ (i.e., a Hadamard transform is applied to both of them), making the state $\frac{1}{2}(|00\rangle + |01\rangle + |10\rangle + |11\rangle)$, this will then naturally evolve into entangled states under the action of $V_{\rm XX}$ alone.

Although we have discussed our Hamiltonian specifically in the context of a dot molecule, these ideas are valid for any nanostructure with corresponding diagonal and off-diagonal interactions where an energy selective excitation is possible. However, we focus on a quantum dot implementation, and in the next sections we shall analyze in detail the interdot and intradot interaction terms and show how they can be tailored as a function of the interdot distance, dot sizes, and material composition.

III. THE MODEL SYSTEM

Different dot geometries (e.g., spherical, pyramidal, or cuboidal shaped dots) can be used to implement the logic



FIG. 4. Schematic diagram of the cuboidal dot model. The cuboids have base sides of lengths 2a and 2b and heights of h_1 and h_2 , respectively. Their centers are separated by a distance *R*. The potential inside the cuboids is set to zero, and that outside them is determined by the band offsets of the conduction and valence bands within the heterostructure.

gates and quantum entanglement schemes discussed above, and in this paper we choose dots of a square-based cuboidal shape (for another possible geometry see Ref. 62). We assume that the potential energy V of both electrons and holes increases abruptly at the cuboid boundaries where the semiconductor band gap changes, and that V=0 inside the cuboids (see Fig. 4). The confinement potential is determined by the band offsets for the electrons and the holes. This type of square-well potential has the advantage of describing both a well-defined dot size in all three dimensions and of bound and unbound solutions in each direction (this is in contrast with, for example, the parabolic potential considered in Ref. 53).

Our model captures the essential properties of quantum dots which are grown by the Stranski-Krastanow (SK) method.⁶³ Such structures show a degree of self-organization⁶⁴ which is ideally suited for the manufacture of prototype quantum devices, and in the realization of the elementary quantum logic gates discussed in this paper. The SK dot growth proceeds through the evaporation of a layer of dot material on to a previously grown substrate: dots form spontaneously due to the competing energy contributions of dot surface area, dot volume, and strain during the growth process. After the formation of dots and subsequent overgrowth of the substrate material, a second layer of dots may be grown; these nucleate preferentially above the dots in the first layer due to the uneven strain field at the surface.⁶⁵ An example of two dots grown in this way is shown in Fig. 5-and it can be seen there that two vertically stacked dots have been grown with a controlled spacing; dots with such characteristics may be well suited for performing the logic



FIG. 5. Transmission electron micrograph of two layers of quantum dots grown by the Stranski-Krastanow method. The dots are made of InAs and the encapsulating material is GaAs; note that the dots in the second layer nucleate preferentially above the dots in the first layer (Ref. 90).

operations described above. As can be seen in Fig. 5, the dots tend to have smaller dimensions in the growth direction than in the perpendicular directions, and the upper dot of the pair tends to be of a slightly larger size, thus allowing for an appropriate identification of the excitation frequencies ω_1 and ω_2 required in our model. Qubit scalability is available via the SK growth procedure since several layers of dots have been shown to grow in stacks.⁶⁶

Our computational Hilbert space requires up to one exciton per dot, and we must therefore calculate both the singleparticle energies, which are determined by the potential profile of the dots, and the two- and four-particle interactions which are determined by the strength of the Coulomb interaction between them. We look at these different quantities separately in this section.

A. Single-particle states

There are a variety of methods for finding the solutions of the Schrödinger equation for electrons or holes in a quantum dot. These include a full pseudopotential calculation,^{67–69} finite element analysis,⁷⁰ plane-wave expansion,⁷¹ and the use of finite differences.⁷² We employ a similar strategy to that in Refs. 73 and 74, where the Schrödinger equation is expanded in a set of analytical basis functions which are the exact solutions of a potential which is close to the one under investigation. This method has the advantage that the state solutions can be stored in a vector in Hilbert space rather than as a wave function amplitude at each of a very large number of different spatial points; time evolution of the quantum states is also easier to simulate when the state is represented by a vector and we shall extend our work to this area elsewhere. Furthermore, a vector representation can allow more physical insight since the basis functions themselves have a known physical interpretation.

Our first step is to express the wave functions for single particles in the envelope function approximation $as^{1,75}$

$$\psi_p(\mathbf{r}) = \phi_p(\mathbf{r}) U_p(\mathbf{r}), \qquad (3)$$

where $\phi_p(\mathbf{r})$ is an envelope function describing the changing wave function amplitude of confined states for particle type *p* over the dot region, and $U_p(\mathbf{r})$ is the Bloch function which has the periodicity of the atomic lattice. In the effective mass approximation, the envelope functions are solutions of the following single-particle Schrödinger equation:

$$\left[-\frac{\hbar^2}{2}\nabla\left(\frac{1}{m_p^*(\mathbf{r})}\right)\nabla + V_p(\mathbf{r})\right]\phi_p^i(\mathbf{r}) = E_p^i\phi_p^i(\mathbf{r}),\qquad(4)$$

where V_p is the confinement potential, which is displayed in Fig. 4 and m_p^* is the effective mass of the particle *p*. These solutions may be obtained by expanding the Hamiltonian in a set of envelope basis functions of the form $\Xi(\mathbf{r}) = \xi_x(x)\xi_y(y)\xi_z(z)$, where the $\xi_i(i)$ are the solutions of a one-dimensional square-well potential with the appropriate effective masses.^{73,74} Both bound and unbound states must be used in the expansion in order to obtain convergent solutions: the forms of these are discussed in the Appendix.

There are two important things to mention about the direct expansion technique. First, the basis functions we have described above do not in general form an orthogonal set if the bound-state solutions have not decayed to zero at the artificial infinite barrier which is used to generate the unbound states. In practice this is rarely a problem, but anyway is circumvented by using a modified basis set which is orthogonal and whose components are linear combinations of the original basis functions. This modified basis set spans the same Hilbert space as the original set. The method we employ to find this set is canonical orthogonalization⁷⁶ which relies on direct diagonalization of the matrix whose elements are the overlap integrals of the basis functions. The Hamiltonian may then be expressed in this new basis as a Hermitian matrix, and solutions are found again by direct diagonalization (we use the NAG diagonalization algorithm in our simulations). The second point is that the basis set must necessarily be truncated; hence the eigenenergies of the solutions we obtain are really upper limits on the true eigenenergies of the coupled dot system (we employ the Rayleigh-Ritz variational method⁷⁷). In practice we can increase the number of basis functions until a sufficiently accurate solution is obtained. We shall mention any important points relating to solution convergence and approximations at the appropriate places later in the paper.

Results of a simple single-particle calculation are displayed in Fig. 6, where the ground-state energy of two particles in QD I (of masses $0.6m_0$ and $0.06m_0$, where m_0 is the free electron mass) is shown as a function of the dot size for two dot geometries. The two geometries correspond to a cubic shape $(a=h_1/2)$ and a flatter cuboidal shape $(a=5h_1)$, which is more typical of SK dots (see Fig. 5). As would be expected, the ground-state energy decreases for a larger dot size and is smaller for heavier particles. The cube shaped dots have ground states with smaller energies than the corresponding cuboidal dot states since the cuboids have one smaller dimension which increases the kinetic energy of the wave function. All of the curves have a kink, and for dot sizes below this the ground-state energy saturates to the value of the confinement potential. This is a consequence of including unbound states in the calculation: once the dots are small enough that the confined state energy has a larger energy than the confinement potential, the ground state becomes unbound and there is hardly any dependence on dot



FIG. 6. The single-particle ground-state energy as a function of dot size for a quantum cube (lower graph) and a quantum cuboid (upper graph). We use the envelope function and effective mass approximations. The filled symbols correspond to $m_p^* = 0.6m_0$ which is typical of heavy holes, and the open symbols correspond to $m_p^* = 0.06m_0$ which is typical of electrons. These values will be used for electrons and holes throughout this paper. The different symbols represent different values for the confinement potential (see legend).

size as it is further reduced. When the ground state is a bound state, the ground energy is always larger in the case of the cuboidal shape, for a given value of *a*: this is a consequence of the smaller height dimension of the cuboid which increases the kinetic energy of the particle. Bound ground states are always very closely approximated by the basis function corresponding to the ground state of the onedimensional well in all three dimensions: in the case of cubic shaped dots, the amplitude of this state is always greater than 0.999; for the cuboidal dot it is always greater than 0.99. We shall use the approximation that the ground state is *exactly* this basis function later in the paper.

B. Coulomb interactions: General methodology

In this section we present our general methodology for calculating the Coulomb interaction matrix elements between electrons and holes in quantum dots. This will be important later for calculations of the intradot and interdot matrix elements in our model system.

First, consider an initial wave function of an *N*-electron system which represents a single exciton state of a quantum $dot:^{68}$

$$\Psi_{\mathrm{I}} = \mathcal{A}[\psi_{1}(\mathbf{r}_{1},\sigma_{1}),\psi_{2}(\mathbf{r}_{2},\sigma_{2}),\ldots,\psi_{s}'(\mathbf{r}_{s},\sigma_{s}),\ldots,\\\psi_{t}(\mathbf{r}_{t},\sigma_{t}),\ldots,\psi_{N}(\mathbf{r}_{N},\sigma_{N})],$$
(5)

where the \mathcal{A} indicates overall antisymmetry (i.e., the wave function takes a Slater determinant form), the σ_i represent the spin state of each electron, and the ψ_i are single-particle wave functions; we have labeled the state *s* with a prime symbol to indicate that it lies in the conduction band, whereas all of the other states are in the valence band.

Next, we assume a final state which is a different single exciton:

$$\Psi_{\mathrm{F}} = \mathcal{A}[\psi_{1}(\mathbf{r}_{1},\sigma_{1}),\psi_{2}(\mathbf{r}_{2},\sigma_{2}),\ldots,\psi_{s}(\mathbf{r}_{s},\\ \times \sigma_{s}),\ldots,\psi_{t}'(\mathbf{r}_{t},\sigma_{t}),\ldots,\psi_{N}(\mathbf{r}_{\mathrm{N}},\sigma_{\mathrm{N}})].$$
(6)

The Coulomb matrix element between these two states is given by

$$M_{\rm IF}^{Coul} = \langle \Psi_{\rm F} | \sum_{ij,i< j} \frac{e^2}{4\pi\epsilon_0\epsilon_r(\mathbf{r}_{\rm ij})|\mathbf{r}_{\rm ij}|} |\Psi_{\rm I}\rangle, \tag{7}$$

where ϵ_0 is the permittivity of vacuum, $\epsilon_r(\mathbf{r})$ is the relative permittivity of the medium (and therefore describes polarization screening), and $\mathbf{r}_{ij} = \mathbf{r}_i - \mathbf{r}_j$. The only nonzero terms in the above expansion are those involving both \mathbf{r}_s and \mathbf{r}_t , since the ground and excited states of each single particle are orthogonal to one another. Hence, we obtain

$$M_{\rm IF}^{Coul} = \langle \Psi_{\rm F} | \frac{e^2}{4 \pi \epsilon_0 \epsilon_r(\mathbf{r}_{\rm st}) |\mathbf{r}_{\rm st}|} | \Psi_{\rm I} \rangle.$$
(8)

Owing to the antisymmetric nature of the wave functions, this matrix element has contributions from a direct term and an exchange term. Both terms take the form of an integral over \mathbf{r}_s and \mathbf{r}_t . In spite of the fact that it arises from the exchanged form of the wave function, the direct term is conventionally written as

$$M_{\rm IF}^{J} = C \int \int \psi_t^{\prime *}(\mathbf{r}_{\rm s}) \psi_s^{\prime}(\mathbf{r}_{\rm s}) \frac{1}{\boldsymbol{\epsilon}_r(\mathbf{r}_{\rm st}) |\mathbf{r}_{\rm st}|} \psi_s^{*}(\mathbf{r}_{\rm t}) \psi_t(\mathbf{r}_{\rm t}) d\mathbf{r}_{\rm s} d\mathbf{r}_{\rm t},$$
(9)

and the exchange term is

$$M_{\rm IF}^{K} = \pm C \int \int \psi_{s}^{*}(\mathbf{r}_{\rm s}) \psi_{s}'(\mathbf{r}_{\rm s}) \frac{1}{\epsilon_{r}(\mathbf{r}_{\rm st})|\mathbf{r}_{\rm st}|} \times \psi_{t}(\mathbf{r}_{\rm t}) \psi_{t}'^{*}(\mathbf{r}_{\rm t}) d\mathbf{r}_{\rm s} d\mathbf{r}_{\rm t}, \qquad (10)$$

where we have introduced the constant $C \equiv (e^2/4\pi\epsilon_0)$. The sign of this exchange term is determined by the spin of the two particles: spin triplet (S=1) states have positive $M_{\rm IF}^K$ elements, whereas spin singlet (S=0) states have negative $M_{\rm IF}^K$ values. We have removed the spin variables from the

single-particle wave functions since these are not important for spatial integrals, once the sign for M_{IF}^{K} has been determined. It is natural now to switch to a hole description of the many-body wave functions introduced above: the matrix elements obtained by the above procedure would be identical if we labeled the states *s* and *t* as holes and simply expressed each wave function as a product of the promoted electron states and the left behind hole states. This allows the other parts of the wave functions to be left out when calculating matrix elements involving these specific electronic states and so makes calculations easier; it also explains the convention on labeling direct and exchange terms. We shall henceforth use such a description.

We proceed further by noting that the two integrals, Eq. (9) and Eq. (10), are both of the form

$$I = \int \int \rho_s(\mathbf{r}_s) f(\mathbf{r}_s - r_t) \rho_t(\mathbf{r}_t) d\mathbf{r}_s d\mathbf{r}_t, \qquad (11)$$

and so we may use Fourier transforms to reduce the dimensionality of the integrand. Employing the convolution theorem and Parseval's relation leads to

$$I = \frac{1}{(2\pi)^6} \int R_s(\mathbf{K}) R_t(\mathbf{K}) F(\mathbf{K}) d\mathbf{K},$$
 (12)

where the Fourier transform of ρ is denoted by *R* and that of *f* is denoted by *F*. We now make the calculation more specific by first assuming that ϵ_r is independent of \mathbf{r}_{st} [see Ref. 68 for a detailed discussion of the form of $\epsilon_r(\mathbf{r}_{st})$], and note that the Fourier transform of the Coulomb operator is given by

$$F(\mathbf{K}) = \frac{4\,\pi}{K^2}.\tag{13}$$

The ρ functions are a product of two wave functions of the form of Eq. (3), and their Fourier transform is simplified by invoking the different length scales of the envelope function and the Bloch function. We may write, for the $M_{\rm IF}^K$ integral (a completely analogous method can be carried out for the $M_{\rm IF}^J$ integrals):

$$R_{s}(\mathbf{K}) = \int \psi_{s}^{*}(\mathbf{r}_{s})\psi_{s}'(\mathbf{r}_{s})\exp(i\mathbf{K}\cdot\mathbf{r}_{s})d\mathbf{r}_{s}$$
$$= V_{cell}\sum_{\mathbf{T}_{i}} \phi_{s,i}^{*}\phi_{s,i}'\exp(i\mathbf{K}\cdot\mathbf{T}_{i})$$
$$\times \int_{cell} U_{s}^{*}(\mathbf{r}_{s})U_{s}'(\mathbf{r}_{s})\exp(i\mathbf{K}\cdot\mathbf{r}_{s})d\mathbf{r}_{s}, \quad (14)$$

where we have assumed that each envelope function takes a constant value ϕ_i over each unit cell *i* of the lattice, that the translational lattice vector of cell *i* is \mathbf{T}_i and that the volume of the unit cell is V_{cell} . We may express the **K** wave vector as

$$\mathbf{K} = \mathbf{k} + \mathbf{G},\tag{15}$$

where \mathbf{k} is a vector within the first Brillouin zone and \mathbf{G} is a reciprocal lattice vector, and we may further convert the sum to an integral to obtain

$$R_{s}(\mathbf{k},\mathbf{G}) = \int_{space} \phi_{s}^{*}(\mathbf{r}) \phi_{s}'(\mathbf{r}) \exp(i\mathbf{k}\cdot\mathbf{r}) d\mathbf{r}$$
$$\times \int_{cell} U_{s}^{*}(\mathbf{r}_{s}) U_{s}'(\mathbf{r}_{s}) \exp[i(\mathbf{k}+\mathbf{G})\cdot r_{s}] d\mathbf{r}_{s}.$$
(16)

Thus R_s is a product of Fourier transforms, one for the envelope functions (which is independent of G, so only needs to be calculated within the first Brillouin zone) and the other for the Bloch functions. It is also obvious that an analogous expression exists for R_t . The envelope function parts of the Fourier transform are analytical: the wave function takes either a sinusoidal or exponential form depending on whether it exists within or outside the quantum dot. We do not write out these expressions explicitly here however, since they are somewhat lengthy and tedious. The Bloch function part is also analytical for a suitable choice of wave function, and to simplify the calculations here we use a Kronig-Penney model where the atomic wave functions are assumed to take the form of the solutions of an infinite square-well potential of well width 2x. Specifically, we assume that the hole states we consider (at the top of the valence band) take the wave function solution of this potential which has p_{τ} symmetry and the electron states we consider take the solution with ssymmetry. We expect this to be sufficient approximation for elucidating the general properties of the system, though a more refined calculation would be required to obtain more accurate estimates of the various quantities we calculate.

By inserting Eq. (16) (and the analogous expression for R_t) and Eq. (13) into Eq. (10) we obtain an expression for M_{IF}^K . The expression has an integrand which is analytical but the integration over three-dimensional **K** space must be carried out numerically. This is done by employing a NAG library routine for multidimensional adaptive quadrature. If the wave function labels are swapped around, an analogous method for calculating M_{IF}^J can be carried out [see Eq. (9)]. We shall show the results of such calculations for various different cases in the next sections.

IV. INTRADOT COUPLING

In this section, we describe the predictions of the above model when it is applied specifically to the calculation of the diagonal matrix element of the two ground-state basis functions representing an electron and a hole on the same dot. In this case the states *s* and *t* are identical, and hence the expression for $M_{\rm IF}^J$ reduces to the direct Coulomb interaction between the ground basis state electron and the ground basis state hole (we call this M_{00}^J). We saw in the preceding section that the ground basis state is a good approximation to the true ground state of the system when only the singleparticle contributions to the Hamiltonian are taken into account. Thus this matrix element is a first-order correction to the energy due to the Coulomb force between the two par-



FIG. 7. The intradot direct Coulomb interaction strength integrand, plotted in the K_x - K_z plane of reciprocal space, and around the **K**=0 point. We have calculated the integrand for dot I and used the cubic geometry ($a=h_1/2=10$ nm), and assumed that $V_e = V_h$ = 500 meV.

ticles. Furthermore, the expression for M_{IF}^{K} reduces to the spin splitting between singlet and triplet exciton states in this first-order approximation.

Let us first consider the direct Coulomb interaction in dot I, which has basal side length a, and simplify things by calculating for a cubic shape (i.e., we set $2a = h_1$). The results may then be directly carried over to dot II, with $a \rightarrow b$ and $2b = h_2$. As we described earlier, it is necessary to evaluate the integral of Eq. (9) by first transforming to reciprocal space and then integrating over K space. The resultant integrand has peaks at each reciprocal lattice point (where k =0 and so the envelope function part of the integrand has a maximum). These peaks quickly die away over a length scale $\sim 1/a$ as would be expected for envelope functions representing wave functions within QD's of side length a. However, it turns out that only the central (G=0) peak is important, since the other peaks contribute much less to the total integral (this is caused by both the k dependence of the atomic contribution to the integrand and by the $1/k^2$ dependence of the Coulomb interaction part). The central peak is displayed in Fig. 7, as a function of K_x and K_y ($K_z=0$).

By numerically integrating the central peak for a range of dot sizes and confinement potentials, we can obtain a plot of the dependence of M_{00}^J on these parameters (this is shown in Fig. 8 for both the cubic geometry and for a flat cuboid, in which $a=5h_1$). As would be expected, the interaction decreases as the size of the QD increases (and so the electron and hole are not forced to be so close together). It is interesting to look at what happens at shorter distances when the confinement potential changes; a larger confinement potential causes a larger Coulomb binding energy. This result is expected since the wave function of both the electron and the resultant closer proximity of the two wave functions causes a larger Coulomb interaction. As would be expected



FIG. 8. The intradot direct Coulomb interaction strength M'_{00} as a function of dot size and confinement potential. The solid lines are for a cubic geometry $(a = h_1/2)$ and the dotted lines are for a cuboid $(a = 5h_1)$.

intuitively, at very large dot sizes the size of the Coulomb interaction scales like 1/a. At very small dot sizes, the direct Coulomb interaction does not follow these simple rules: for a weak enough confinement there is a peak in the energy and at small values of a it decreases. This can be understood by thinking about the shapes of the wave functions in this region. When the well width is small, the curvature of the wave function is necessarily rather high, and so the kinetic energy is large. In order to compensate for this, the wave function spreads out into the barriers at the cost of some potential energy, if the barriers are not too high (this energy cost is balanced by the saving in kinetic energy). Thus the wave function has a larger size than would naively be expected from the dot size, and the Coulomb binding energy decreases. The peak in the Coulomb potential occurs for a larger value of a in the cases of cuboidal geometry: this is simply because of the shorter height dimension in this case, which means that the wave function spread effect discussed above remains significant at larger values of a. At still larger values of a, the Coulomb interaction is larger for the cuboidal geometry than it is for the cube: again this is because the cuboid has one smaller dimension, which means that the electron and hole are forced to be closer together in the cuboidal case.

We next look at the value of the exchange coupling between the two ground-state basis functions (i.e., the spin singlet-triplet splitting M_{00}^K , correct to first order). The relevant *K*-dependent integrand takes a somewhat different form in this case. The central peak (around **G**=0) is displayed in Fig. 9, where it can be seen that the function has a zero at K=0; this is expected since the electron and hole wave functions have opposite parity. The suppression at K= 0 means that, this time, the regions around other reciprocal lattice points have to be included in the numerical integration. The resultant dependence of M_{00}^K on dot size and confinement potential is displayed in Fig. 10.



FIG. 9. The intradot exchange Coulomb interaction strength integrand, plotted in the K_x - K_z plane of reciprocal space, and around the **K**=0 point. Note that it has value zero at **K**=0 and is asymmetric in *x* and *z* due to the choice of the p_z Kronig-Penney state for the holes. We have calculated the integrand for dot I and used the cubic geometry ($a=h_1/2=10$ nm) and assumed that $V_e=V_h$ = 500 meV.

By reference to Fig. 10 we see that the exchange splitting is several orders of magnitude smaller than the direct Coulomb term, though it follows the same trends of increasing in value with smaller dot sizes and larger confinement potentials. These effects can be understood as follows: the exchange splitting is essentially a consequence of Pauli's exclusion principle which states that particles in the same quantum state cannot exist together at the same spatial point. Thus electrons and holes which are in a triplet spin state (and so have indistinguishable spin properties) necessarily "avoid" each other, thus reducing the Coulomb attraction



FIG. 10. The intradot exchange Coulomb interaction strength M_{00}^{K} as a function of dot size and confinement potential. The solid lines are for a cubic geometry $(a=h_1/2)$ and the dotted lines are for a cuboid $(a=5h_1)$.

between them. This effect is expected to be more significant when the wave functions in the absence of the Coulomb interaction overlap strongly-that is when they are localized in one small region of space. There is, in general, a greater degree of localization of the wave functions when either the dots are smaller or when the confinement is stronger, and hence the exchange splitting gets larger when these conditions are satisfied. However, there is again one region of Fig. 10 where this general rule is not obeyed-that is, when confinement is relatively weak, but where the dot size is small. Here, the exchange energy takes a downturn as the dot size gets smaller. This is caused by increased wave function barrier penetration, which again means the effective wave function size gets *larger* rather than smaller as might be expected from the simple intuitive picture described above. The comparison between the cubic and cuboidal geometries shows a similar trend to that discussed above for the case of the direct Coulomb interaction, and the reasons for this follow the same lines. The turnover occurs for a larger *a* for the cuboidal dot due to the wave function spreading effect which occurs for smaller spatial dimensions: this spreading causes a reduction in the exchange splitting. At larger sizes, the exchange splitting is larger for the cuboidal shape due to the effect of the smaller height dimension of the cuboid, which pushes electron and hole together and thus increases the exchange splitting. At larger dot sizes, the interaction scales like $1/a^3$.

V. INTERDOT INTERACTIONS

In this section we shall discuss the interdot coupling terms which are due to the Coulomb operator introduced in previous sections. These terms are crucial to the operation of a quantum device, since they may allow qubit-qubit interactions to take place, which is an essential requirement for two (or more) qubit gates to be constructed. There are two important types of interaction which may occur. The first type is called the Förster interaction, and is described by an offdiagonal matrix element (in the computational basis) between two single exciton wave functions of the type introduced in Eqs. (5) and (6), but where the two excitons are located on different quantum dots this interaction is called $V_{\rm F}$ in Eq. (1)]. The second interaction which is important for this scheme is the direct self-Coulomb interaction in a biexciton (double substitutional Slater determinant) wave function, where one exciton is located on each dot. This is called $V_{\rm XX}$ in Eq. (1) and amounts to the Coulomb binding energy between two excitons located on adjacent dots. We next quantify both of these interaction terms and discuss their properties within the context of the quantum computing implementation described in Sec. II.

A. Off-diagonal coupling: Förster interaction

The Förster or Coulomb exchange interaction can induce the transfer of an exciton from one quantum dot to the other. This is a nonradiative energy transfer whereby an exciton is destroyed on one dot and recreated on the other; it is an electrostatic interaction which proceeds via a shortlived virtual photon. Förster's original theory⁷⁸ showed that the inter-



FIG. 11. The plot of the integrand in **K** space which leads to the Förster strength, as a function of K_x and K_z ($K_y=0$). The extra modulation on the function (as compared with Figs. 7 and 9) is caused by the interdot separation. The plot is for cubic dots with a=b=10 nm and R=20 nm, and we have taken $V_e=V_h$ = 500 meV.

action is dipole-dipole to lowest order; this theory was subsequently elaborated by Dexter⁵⁵ who derived higher-order and exchange terms in studies of the sensitized luminescence of solids. Here we extend this theory to the case of the manybody exciton states of quantum dots. The off-diagonal nature of the interaction causes the eigenstates of the Hamiltonian, Eq. (1), to become linear combinations of the computational basis states $|10\rangle$ and $|01\rangle$. As described in Sec. II, the degree of this mixing is crucial in determining how to generate quantum entanglement in the quantum dot molecule. The Förster coupling can be expressed as the matrix element of the direct Coulomb operator between excitons located on each of the two dots:

$$V_{\rm F} = C \int \int \psi_s(\mathbf{r}_{\rm s}) \psi_s'(\mathbf{r}_{\rm s}) \frac{1}{\epsilon_r (\mathbf{R} + \mathbf{r}_{\rm s} - r_{\rm t}) |\mathbf{R} + \mathbf{r}_{\rm s} - r_{\rm t}|} \\ \times \psi_t(\mathbf{r}_{\rm t}) \psi_t'(\mathbf{r}_{\rm t}) d\mathbf{r}_{\rm s} d\mathbf{r}_{\rm t}.$$
(17)

This equation is equivalent to Eq. (9), but we have explicitly included the interdot vector R in the Coulomb operator; we assume in this case that the two variables \mathbf{r}_{s} and \mathbf{r}_{t} are defined from the centers of dot I and dot II, respectively. We may evaluate $V_{\rm F}$ in exactly the same way as we evaluated the intradot couplings, so long as the new positions of the wave functions are included in the calculation. An example of the integrand appearing in Eq. (17) is shown in Fig. 11. It is interesting to compare this figure to Figs. 7 and 9; in the plot of Fig. 11 there is an extra modulation due to the extra factor associated with the interdot separation, and this added modulation means the integral takes longer to evaluate numerically. The results are displayed in Figs. 12 and 13, where the Förster strength is displayed as a function of dot separation, shape, and confining potential. The data are displayed on a log scale, and it can be seen that, for the cubic shape, they



FIG. 12. Dependence of the Förster interaction strength on the interdot separation *R*. The dots have equal sizes and results are shown for two shapes: (i) cubic, with a=h/2=2 nm (upper two curves), and (ii) cuboidal, with a=5h=2 nm (lower two curves). The circles and squares represent the predictions of a full numerical simulation for well depths of 500 meV, and 2000 meV respectively (the electron and hole wells are assumed to be of the same depth). The dotted lines represent the predictions of the dipole-dipole model for the cubic shaped dots and the dashed curve represents the maximum coupling predicted for the dipole-dipole model, which corresponds to $O_{\rm I}=O_{\rm II}=1.0$. The dotted lines can be obtained by multiplying this maximum by the relevant values of $O_{\rm i}$ which can be obtained from Fig. 14.

closely follow a $1/R^3$ law for all the separations considered. This form is expected for a dipole-dipole interaction, and we shall now discuss how a power-series expansion and subsequent approximation lead to this type of interaction in this case. In so doing, we shall also explain why the interaction is modified as the size and shape of the dots are changed.

By making the assumption that **R** is much larger than \mathbf{r}_s and \mathbf{r}_t , we may Taylor expand the Coulomb operator. This procedure yields, to lowest nonzero order,

$$V_{\rm F} = \frac{C}{\epsilon_r R^3} \left(\langle \mathbf{r}_{\rm I} \rangle \cdot \langle \mathbf{r}_{\rm II} \rangle - \frac{3}{R^2} (\langle \mathbf{r}_{\rm I} \rangle \cdot \mathbf{R}) (\langle \mathbf{r}_{\rm II} \rangle \cdot \mathbf{R}) \right), \quad (18)$$

where it has been assumed that the dielectric constant is independent of $\mathbf{R} + \mathbf{r}_s - \mathbf{r}_t$, and as throughout the paper is assumed to take the constant value of $\epsilon_r = 10$. The matrix element of the position operator between an electron and a hole state on dot I or II is

$$\langle \mathbf{r}_{I/II} \rangle = \int \psi'_{I/II}(\mathbf{r}) \mathbf{r} \psi_{I/II}(\mathbf{r}) d\mathbf{r}.$$
 (19)

Equation (18) is therefore equivalent to the interaction of two point dipoles, one situated on each dot. We can proceed further by again employing the envelope function approximation for electrons and holes [Eq. (3)] and by rewriting Eq. (19) as



FIG. 13. Dependence of the Förster interaction strength on the shape of the dots. The dots are assumed to be identical, but have a series of cuboidal shapes with different aspect ratios. The well depth is 500 meV (the electron and hole wells are assumed to be of the same depth). The solid curve represents the maximum coupling predicted for the dipole-dipole model, which corresponds to $O_1 = O_{II} = 1.0$.

$$\langle \mathbf{r}_{I/II} \rangle = \sum_{\{\mathbf{T}_i\}} V_{cell} \int_{cell} \phi'_{I/II} (\mathbf{r} - \mathbf{T}_i) U'(\mathbf{r}) (\mathbf{r} - \mathbf{T}_i) \times \phi_{I/II} (\mathbf{r} - \mathbf{T}_i) U(\mathbf{r}) d\mathbf{r},$$
(20)

where T_i represents the set of lattice vectors. We have made use of the periodicity of the Bloch part of the wave functions and assumed that this part of the wave function is the same for both dots. By making the assumption that the envelope function is slowly varying on the length scale of the atomic lattice and by using the orthogonality of the electron and hole Bloch functions we find that

$$V_{\rm F} = \frac{C}{\epsilon_r R^3} O_{\rm I} O_{\rm II} \bigg(\left(|\langle \mathbf{r}_{\rm a} \rangle|^2 \right) - \frac{3}{R^2} (\langle \mathbf{r}_{\rm a} \rangle \cdot \mathbf{R})^2 \bigg), \qquad (21)$$

where the term $\langle \mathbf{r}_a \rangle$ represents the atomic position operator expectation value

$$\langle \mathbf{r}_{a} \rangle = \int_{cell} U_{e}(\mathbf{r}) \mathbf{r} U_{h}(\mathbf{r}) d\mathbf{r},$$
 (22)

which is the same for both dots and

$$O_{i} = \int_{space} \phi_{e}^{i}(\mathbf{r}) \phi_{h}^{i}(\mathbf{r}) d\mathbf{r}$$
(23)

is the overlap of electron and hole envelope functions on the appropriate dot i. Equation (21) shows how the effects of the quantum dot size and shape (which determine the overlap integrals) may be separated from the effects of the material composition of the dot (which determine the atomic dipole operator).



FIG. 14. The overlap integral O_i [Eq. (23)] as a function of dot size and confinement potential. This graph can be used in conjunction with Fig. 12 to obtain values of the Förster strength for a range of dot sizes and confinement potentials.

It is now possible to obtain the strength of the interaction by assuming the specific forms for the envelope and atomic functions which we discussed earlier, in Sec. III. For the Kronig-Penney model, with a well width of 2x, the atomic position operator expectation value is given by

$$\langle \mathbf{r}_{\mathrm{a}} \rangle = 32x/9\pi^2. \tag{24}$$

The overlap integrals are easily calculated for the envelope functions described earlier, and are displayed as a function of dot size and confinement potential in Fig. 14. We show the overlap integral for the usual two dot shapes: cubic (a =h/2) and flat cuboidal (a=5h), where the latter is more typical of dots grown by the Stranski-Krastanow method. The overlap is enhanced when there is a larger confinement potential and for larger dots, since in these cases the shape of the wave function is less sensitive to the effective mass difference of the electrons and holes. We may use Fig. 14, together with the atomic dipole value, to calculate the Förster strength for a range of dot sizes and confining potentials. Owing to its dependence on the atomic dipole operator ($V_{\rm E}$ $\propto x^2$), we plot V_F/x^2 as a function of R in Fig. 12. Two example curves, for two equally sized cubic dots are shown for equal electron and hole potentials of 500 meV and 2000 meV in Fig. 12, together with the earlier full calculation in which the dipole-dipole approximation was not made. The full calculation was carried out for (i) cubic dots with a=h/2=2 nm, and (ii) cuboidal dots with a=5h=2 nm. For both shapes it is clear that the influence of dot shape and size is much more important in determining the size of the interaction than the influence of the size of the confinement potentials. Furthermore, the dipole-dipole approximation is very good in the case of a cubic dot, even at interdot separations which are relatively small when compared to the dot sizes. For the cuboidal dot, the dipole-dipole approximation fails at smaller separations. We investigated this effect more thoroughly by repeating the calculations for cuboids of different aspect ratio (see Fig. 13). The next-order term (dipole quadrupole) is zero in all cases when the dots have equal size; presumably the reason for the accuracy of the dipoledipole approximation in the case of cubic dots is that the dipole-dipole terms dominate the higher-order terms even at smaller dot separations.

The simple Kronig-Penney model shows how the size of the Förster transfer depends upon the physical size of the atomic part of the wave function. However, $\langle \mathbf{r}^a \rangle$ is a widely measured quantity since it determines the strength of dipole allowed transitions in optical spectra. In CdSe QD's it can be in the range of 0.9 to 5.2*e* Å,⁵⁷ in atomic systems it can also be several *e* Å (Ref. 79) and in BS's and MS's has recently been observed to be about 1.7 *e*Å.⁵⁸

As an illustration of the use of these curves, let us assume that we have a dot system in which, as before, R = 5 nm, a = 10 nm, b=8 nm, and $h_1=h_2=2$ nm. Furthermore, let us take the measured dipole value for CdSe dots of 0.9 to 5.2 eÅ.⁵⁷ In this case, the Förster strength is between 0.013 and 0.45 meV, which if $\Delta_0 = 0$ would correspond to an on resonance energy transfer time of between 318 and 9.2 ps in dots with $V_h = V_e = 500$ meV. This is short enough to be useful for quantum computing purposes: decoherence times as long as a few nanoseconds⁸⁰ have been observed in QD's. In MS's or BS's, the interacting units can be as close together as 1 nm; using this and taking a typical molecular or biomolecular dipole value of about 1.7e Å, 57,58 we obtain an interaction strength of 8.3 meV (or a transfer time of \sim 497 fs). Furthermore, V_F must certainly be controlled if the alternative scheme using V_{XX} is to be implemented (and therefore cannot be neglected as in Ref. 53). We note that $V_{\rm F}$ is not particularly sensitive to differences in dot size, though the differences in the diagonal (self-energy) parts of the Hamiltonian which are caused by having dots of unequal size are very significant. We shall discuss this further in Sec. VI.

In Sec. V B we shall discuss how the biexciton bindingenergy term depends upon applied electric field, and how such a field may be crucial to the operation of a potential quantum logic device. We now discuss how the Förster term would vary when such a field is applied. Since an electric field would move the electron and hole away from one another, the overlap integrals, Eq. (23), would be reduced by such a field. We show this specifically by simulating the effect of applying a field on the overlap integral O_i in Figs. 15 and 16. In Fig. 15, we see that O_i is significantly suppressed for fields of a few tens of kV/cm, and that the suppression is easier to achieve in dots which have a larger dimension in the direction of the applied field. The reason for this is that in larger dots the electron and hole are more easily separated since there is more distance between the two dotbarrier interfaces. In zero field, dots with a smaller dimension have smaller overlap integrals, for the reasons associated with the balance of kinetic and potential energy discussed earlier. Hence, we see that the curves for different dot sizes cross each other in an applied field. In Fig. 16 we plot the dependence of the O_i on the depth of the confinement potentials, for a = 10 nm for both the cubic and cuboidal geometries. We see here that the effect of varying confinement potential is much smaller than varying dot size. The small difference that is evident, that of a slightly easier suppression for deeper confinement potentials, is presumably



FIG. 15. The overlap integral O_i [Eq. (23)] as a function of electric-field strength *E* for a range of dot sizes. The upper part shows the dependence when the dots take a cubic shape $(a = h_1/2)$; the lower part shows the dependence for a cuboidal shape $(a=5h_1)$. Note that the overlap integral, and so also the Förster interaction, is suppressed at large field as the electron and hole are forced apart.

due to the fact that the wave functions in a shallower wells tend to be more spread out (since the potential-energy cost in doing so is smaller), and so the overlap between an electron and hole at opposite sides of the dot is slightly enhanced. Again the curves cross one another since, as we discussed earlier, in zero field the delocalization of the states for shallower potentials means that the shape of each particle's wave function depends more strongly on its effective mass.

The fact that the Förster coupling may be suppressed by an external field could be very useful: if an entangled state is produced by using this coupling, it may be maintained by switching off $V_{\rm F}$. If this could be done in a sufficiently short time (i.e., on the time scale of the evolution of the quantum device under the Förster coupling Hamiltonian, but much less than typical decoherence times), it may be possible to fabricate a two-qubit gate using this effect. It may also be possible to achieve this switching in an alternative way, by leaving the Förster coupling at a constant value but by tuning the single exciton level spacing Δ_0 through the electric field



FIG. 16. The overlap integral O_i [Eq. (23)] as a function of electric-field strength *E* for a range of confinement potentials, for a dot size of a=10 nm. The upper part shows the dependence when the dots take a cubic shape $(a=h_1/2)$; the lower part shows the dependence for a cuboidal shape $(a=5h_1)$.

which induces the Stark shift;¹ we shall return to this in Sec. VI. It is also interesting that a negligible Förster coupling is essential for the energy selective dot device discussed in Sec. II: such a negligible coupling may be achieved through using an external field. There are disadvantages doing this in this case, however. A smaller electron-hole overlap will also reduce the coupling to the light field itself, which we need to be strong enough to be able to perform conditional gates in a time short enough when compared with typical decoherence times. Hence, as so often in quantum computing implementation, a compromise must be struck between these two requirements.

B. Diagonal coupling

We now calculate the direct Coulomb interaction between two excitons, where one exciton is located on each dot. This interaction leads to the energy selectivity of the gate and is responsible for the V_{XX} term of Eq. (1).

OPTICAL SCHEMES FOR QUANTUM COMPUTATION IN ...

Consider the following double substitutional Slater determinant, which we write using the hole prescription described in Sec. III B. It represents a combination of two ground conduction electron states and two ground hole states which correspond to one exciton on each dot:

$$\Psi_{\mathrm{XX}} = \mathcal{A}[\psi_{\mathrm{e}}^{\mathrm{I}}(\mathbf{r}_{1})\psi_{\mathrm{h}}^{\mathrm{I}}(\mathbf{r}_{2})\psi_{\mathrm{e}}^{\mathrm{II}}(\mathbf{r}_{3}-\mathbf{R})\psi_{\mathrm{h}}^{\mathrm{II}}(\mathbf{r}_{4}-\mathbf{R})], \quad (25)$$

where \mathcal{A} indicates that the wave function has overall antisymmetry, this being achieved by adding terms with labels swapped around in a Slater determinant form. **R** is the vector connecting the two dot centers, **r**₁ and **r**₃ represent the position vectors of electrons relative to the centers of dot I and dot II, respectively, and **r**₂ and **r**₄ are the equivalent vectors for holes. The associated Coulomb operator \hat{V}_{XX} is given by

$$\hat{V}_{XX} = \frac{C}{\epsilon_r} \left[\frac{1}{|\mathbf{R} + \mathbf{r}_1 - \mathbf{r}_3|} - \frac{1}{|\mathbf{R} + \mathbf{r}_1 - \mathbf{r}_4|} - \frac{1}{|\mathbf{R} + \mathbf{r}_2 - \mathbf{r}_3|} + \frac{1}{|\mathbf{R} + \mathbf{r}_2 - \mathbf{r}_4|} \right].$$
(26)

Expanding this expression in a Taylor series about ${\bf R}$ gives, to lowest nonzero order,

$$\hat{V}_{XX} = \frac{C}{\epsilon_r R^3} \left\{ \mathbf{p}_{I} \cdot \mathbf{p}_{II} - \frac{3}{R^2} (\mathbf{p}_{I} \cdot \mathbf{R}) (\mathbf{p}_{II} \cdot \mathbf{R}) \right\}, \qquad (27)$$

where $\mathbf{p}_{I} = e(\mathbf{r}_{1} - \mathbf{r}_{2})$ is the overall dipole moment on dot I and $\mathbf{p}_{II} = e(\mathbf{r}_{3} - \mathbf{r}_{4})$ is the overall dipole moment on dot II. To evaluate the matrix element $\langle \Psi_{XX} | \hat{V}_{XX} | \Psi_{XX} \rangle$, \mathbf{p}_{I} and \mathbf{p}_{II} in Eq. (27) are replaced by their expectation values for the wave function, Eq. (25). This procedure gives rise to a direct term and exchange terms. The exchange terms arise from the parts of the wave function [Eq. (25)] which do not appear explicitly within the bracket but which have their labels swapped around and they are zero in the absence of wave function overlap between dots. The direct term is obtained through the use of the envelope function approximation, Eq. (3), which leads to the following equation for the expectation value $\langle \mathbf{r}_{1} \rangle$:

$$\langle \mathbf{r}_1 \rangle = \int_{space} \phi_{e}^{I*}(\mathbf{r}_1) \mathbf{r}_1 \phi_{e}^{I}(\mathbf{r}_1) d\mathbf{r}_1, \qquad (28)$$

where the orthogonality of the Bloch functions for different bands and the slow variation approximation for the envelope functions have again been used. Similar expressions hold for the other position expectation values.

For a cubic dot, where the electron and hole wave functions have a definite parity about the dot center, Eq. (28) implies that the exciton-exciton coupling is zero. However, this is not the case when this symmetry is broken. For instance, for pyramidal shapes the electron may localize in one region of the dot and the hole in another region.⁷² Alternatively, an electric field would induce a polarization on the dot; this field may be externally applied or arise from intrinsic piezoelectric effects.^{47,53}

We have simulated the effect of applying an electric field in our cuboidal model by including a linear potential in the



FIG. 17. (a) Exciton-exciton binding energy and (b) induced dipole moment as a function of the dot size, shape, and applied electric field. We have assumed that $V_e = V_h = 500$ meV.

single-particle Schrödinger equations (4), and the results are displayed in Fig. 17. In the lower part of the figure we display the size of the exciton dipole moment p_i on one of the dots as a function of the dot size and of the applied electric field strength E. We do this for our usual two geometries: cubic $(a=h_1/2)$ and flat cuboid $(a=5h_1)$ (see Fig. 4). In both cases the field is applied along one of the axes of the square base. The interaction strength V_{XX} is then obtained by using the size of p_i for each dot and substituting into Eq. (27). Thus, the upper part of Fig. 17 shows the strength V_{XX} , normalized by R^3 , for two dots of equal size and calculated for both of the dot geometries just described. [The interaction between two unequally sized dots can similarly be obtained by the use of Fig. 17(b) and Eq. (27).] At very small applied field, the induced dipole is linearly proportional to the field, and hence the interaction strength takes a quadratic dependence on field. At larger applied fields, the induced dipole begins to saturate as the electron and hole approach the edges of confining potential of the quantum dots; this limits the useful interaction strength which may be obtained from a given pair of dots. It is interesting that the interaction strength is much more dependent on a than it is on h_1 ; this is because it is in the basal plane that the field is applied, and so it is in this direction that the dipole moment is induced. The relative insensitivity of V_{XX} to h_1 turns out to be very useful: it means that SK dots, which can be stacked closely on top of one another but which have a relatively large base size, can be made to interact very strongly. As an illustration, consider the following typical SK dot parameters: R=5 nm, a = 8 nm, b=10 nm, and $h_1=h_2=2$ nm (we have assumed the usual experimental situation in which the upper dot of the stack has a slightly larger size). In an *x* directed field of 100 kV/cm, these parameters give $V_{XX} \approx 120$ meV, which would result in a lower time limit for the gate operation of around 10 fs. This is relatively short; decoherence times on the order of nanoseconds have been observed for uncoupled dots.⁸⁰ Finally, we note that we have only calculated V_{XX} to first order; in some cases higher-order terms may be important.

VI. FURTHER DISCUSSION

The model outlined above can also be used to calculate Δ_0 , the difference in exciton creation energy for two different sized dots in the absence of interactions. This is done by simply calculating the single-particle electron and hole energies and taking into account the Coulomb binding energy between them. We also assume that the electron and hole are in the spin singlet state; spin is not important for our present proposal and it is always possible to choose the spin singlet state by using light with the appropriate polarization. Hence we can effectively ignore the triplet states in considering the scheme described in Sec. II.

The absolute value of Δ_0 is displayed as a function of the ratio of the dot side lengths for each of the usual geometries in Fig. 18. We have displayed Δ_0 , both in the absence of the (intradot) Coulomb terms, and when these terms are included to first order by using the calculations of Sec. IV. Δ_0 is zero when the dots are of equal size, and then increases as the difference in size becomes greater. The Coulomb terms serve to reduce the size of Δ_0 because they are larger for the smaller dot of the pair (which also has the larger single-particle energies).

We now use the analysis of Sec. II and the calculations of subsequent sections to obtain the size of the c_1 component of the $|\Psi_{10}\rangle$ and $|\Psi_{01}\rangle$ states. If we assume that the Förster strength is small in comparison with Δ_0 , then we have that $c_1 \approx V_F / \Delta_0$ [see inset of Fig. 1(c)] and by substituting Eq. (24) and Eq. (21), we obtain

$$\frac{R^3 c_1}{x^2 O_1 O_{\rm H}} \approx \frac{37.1}{\Delta_0},\tag{29}$$

where Δ_0 is measured in meV, and *x* and *R* are in nm. This quantity is displayed as a function of dot size ratio for the usual geometries in Fig. 19. It can be seen there that a range of c_1 values can be obtained by choosing dots with appropriate values of *x*, *R*, and *a/b*. For example, cubic dots with large *x* (>1 nm say), small *R* (<3 nm say), and *a/b*~1 give a larger c_1 , and it is then more appropriate to use the Förster interaction itself to create entangled states. On the other hand, dots with smaller *x*, larger *R*, or a large mismatch in dot size would be more suited to the scheme which uses the V_{XX} for QC and entanglement generation. A scheme similar to the latter one was discussed by Biolatti *et al.* in Ref. 53, though the off-diagonal coupling was not considered there at all; we now see how important it is to consider the effect of this interaction. The *fidelity*⁸¹ of a typical V_{XX} en-



FIG. 18. Energy splitting $\Delta_0 \equiv \omega_1 - \omega_2$ of the singlet qubit exciton states $|\Psi_{01}\rangle$ and $|\Psi_{10}\rangle$ for (a) cubic dots $(a=h_1/2,b=h_2/2)$ and (b) cuboidal dots $(a=5h_1,b=5h_2)$ in the absence of the Förster interaction as a function of the dot size ratio a/b. The splitting is independent of interdot distance. The solid lines represent the splitting in the presence of the Coulomb and exchange splitting terms and each adjacent dotted line represents the splitting without Coulomb interactions for each dot size.

tangling gate operation (e.g., $|11\rangle \mapsto |10\rangle$) is equal to $1-c_1^2$ —and so one must be careful when using the biexciton scheme to use the available parameter space and make sure that the Förster transfer is suppressed to the desired accuracy. There are other sources of decoherence in this case (e.g., the interaction with optical and acoustic phonons^{9,61,80}) which will reduce the value of the fidelity to below $1 - c_1^2$. To minimize the effects of such decoherence channels, it is important to maximize V_{XX} , since this leads to an improved transition discrimination and so to a faster gating time. This can be done by applying an electric field and choosing an appropriate dot shape, size, and separation (as described earlier). It is then necessary to minimize the basis state mixing for the chosen parameters by selecting a suitable dot size ratio and material composition. It was seen earlier that the value of $V_{\rm XX}$ could be as high as several tens of meV. If we assume a conservative figure of 10 meV, we find that the uncertainty principle implies that a controlled-NOT gate could be per-



FIG. 19. The size of the component of the wave function c_1 for (a) cubic dots $(a=h_1/2,b=h_2/2)$ and (b) cuboidal dots $(a=5h_1,b=5h_2)$ as a function of the dot size ratio a/b. c_1 has been scaled by its dependence on the interdot distance *R*, typical atomic spacing *x*, and overlap integrals O_i .

formed in a time of around 100 fs. This is relatively short; decoherence times on the order of nanoseconds have been observed recently for uncoupled dots.⁸⁰ Hence we conclude that this scheme looks rather promising as a solid-state implementation of quantum computation.

It is also possible to use the Stark effect¹ to tune two nonresonant levels into resonance, thereby allowing for the kind of switching of the Förster interaction which is required if it is to be used for quantum logic. This can be done so long as the two dots are made such that they have different polarizabilities (which could be achieved by using either different sized dots or dots made of different materials). So long as this difference is such that the levels are brought closer together by applying a field, and that shifts as large as Δ_0 can be achieved, switchable resonant transfer is possible. It might be difficult to achieve the switching in a time which is short enough for a quantum gate to be performed; however, in this case the optical (ac) Stark shift could be employed by using ultrafast lasers.⁸²

Single shot qubit state measurement in QD's could be performed by using resonant fluorescent shelving

techniques.⁸³ The OD state measurement can also be achieved by means of projecting onto the computational basis and measuring the final register state by exploiting ultrafast near-field optical spectroscopy and microscopy:60,84 these allow one to address, to excite, and to probe the QD excitonic states with spectral and spatial selectivity. In addition, the qubit register density matrix can be reconstructed by measuring the QD photon correlations via standard quantum state tomography techniques.⁸⁵ In particular, we believe that the activity of the (Förster) resonant energy transfer processes discussed in this paper can be accomplished in our coupled dot molecule by measuring the intensity correlation function (usually denoted $g^{(2)}$) in a Hanbury-Brown/Twiss type experiment.^{6,8,86} Such an experiment can reveal signatures of purely nonclassical photon correlations arising from the QD molecule emission (i.e., photon antibunching or bunching behavior). This idea has already been experimentally explored for the case of pairs of dye molecules by Berglund et al.⁸⁶ Scalability of the scheme given here could also be possible by adopting a globally addressed qubit strategy⁸⁷ on a stack of self-organized QD's.⁶⁴

We finalize this section with a discussion on how the $V_{\rm F}$ coupling can be usefully manipulated in biomolecular nanostructures. Light-harvesting antenna complexes⁵⁶ or arrays of strongly interacting individual molecules⁵⁸ could provide an appropriate system in which the Förster interaction could be used for quantum information processing tasks. They are generally very uniform structures, and we may compare them to QD's by setting $a/b \sim 1$, or $V_{\rm F}/\Delta_0 \gg 1$. Then the one-exciton eigenstates of a two-qubit system with a Förster coupling naturally allow the generation of the states $1/\sqrt{2}(|01\rangle \pm |10\rangle)$, which, apart from their applications to quantum protocols, can be particularly useful in the fight against decoherence. Spectroscopic, line-narrowing techniques (e.g., hole burning and site-selective fluorescence), infrared and Raman experimental studies reveal that the main decoherence mechanisms in the antenna complexes arise from energetic disorder, electron-phonon coupling, and temperature effects.⁵⁶ In this scenario, the excitations couple to an environment that typically possesses a much larger coherence length than the biomolecular units (BChl's) spacing. For example, the BChl's in the antenna complex LH2, which we regard as a potential system for quantum logic, are spaced by as little as 1 nm, and hence so-called collective decoherence is expected to apply. In this case, provided that the logical qubit encoding $|\downarrow\rangle_i \equiv |01\rangle_{jk}$, $|\uparrow\rangle_i \equiv |10\rangle_{jk}$ that uses two physical (exciton) qubits can be realized in the BChl's system, arbitrary superpositions of logical qubits such as $(\alpha_i |\downarrow\rangle_i + \beta_i |\uparrow\rangle_i)^{\otimes N}$, i = 1, ..., N, $\alpha_i, \beta_i \in \mathbb{C}$, are immune to dephasing noise (described by a σ_z operator⁶¹), and single-qubit manipulations can be carried out on the time scale of the Förster coupling (which as we have seen can be as short as 497 fs). Two-qubit logic gates can also be implemented within a decoherence-free subspace by using the above encoding,⁸⁸ thus completing a universal set of gates. Initialization of the system requires the pairing of the physical qubits to the logical "ground" state $|\downarrow\rangle_i^{\otimes N}$, and readout is to be accomplished by identifying on which of the two structures the exciton resides. Furthermore, rings of BChl's appear side by side in naturally occurring antenna complexes and also display energy selectivity—smaller rings tend to have higher energy transitions.⁵⁶ Thus, following a scheme as above, it may be possible to scale up such biological units in a natural way and construct a robust energy selective scheme for quantum computation.⁸⁹ We also note that arrays of strongly interacting individual molecules which are coupled via a near-field dipole-dipole interaction are well suited for our quantum computing and entangling schemes, especially due to the existence of V_{XX} type of energy shifts,⁵⁸ which have been analyzed in this paper. A full discussion of these ideas will be presented elsewhere.⁸⁹

VII. SUMMARY

We have shown that it is possible to use the two different electrostatic coupling terms (Förster transfer and biexciton binding energy) between excitons in quantum dots to construct two qubit gates which, in addition to an appropriate control over single qubits, are enough for universal quantum computation. We have also discussed how to generate tailored exciton entangled states by using these gates. We have furthermore modeled a pair of quantum dots in the simplest envelope function approximation, and have mapped out the areas of parameter space where one of the two interactions dominates. We have discussed in detail how to perform entangling operations when one of the two interactions is dominant; the case when the two have similar magnitude leads to a rich spectrum of entangled states whose degree of entanglement can be quantified a posteriori in, for example, photon correlation experiments. We have concentrated here on two geometries, namely that of a cube-shaped dot and that of a cuboid-shaped dot, and our calculations have been partially analytical and partially numerical. In future, we are hoping to obtain simpler analytical results for many of the quantities we have calculated, by using different dot geometries. We shall also extend our calculations to include decoherence effects, and in future work we shall be particularly interested in how we might use the Förster interaction to create and use decoherence-free subspaces.

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APPENDIX: SINGLE-PARTICLE SOLUTIONS

As we discussed in the text, the solution of the threedimensional finite well square box potential is obtained by using the solutions of the one-dimensional finite square well and expanding the Schrödinger equation in these basis states. In order to do this, both bound and unbound basis states had to be taken into consideration.

1. Bound states

The problem of finding the solutions of a finite square well is covered in most undergraduate text books (see, e.g.,

TABLE I. Table of the forms of the bound-state solutions for the finite square-well potential.

| Region | Even solutions | Odd solutions |
|-------------------------------------|---|---|
| x < -a/2 $-a/2 < x < a/2$ $x > a/2$ | $A e^{\alpha x}$ B cos(kx) A e ^{-\alpha x} | $Ce^{\alpha x}$ D cos(kx) - Ce ^{-\alpha x} |

Ref. 77), and so we shall not go into too much detail here. If we assume that the finite well is centered around x=0, we have different forms of solutions depending on the parity of the wave function. These forms of solutions are shown in Table I with reference to Fig. 20, and assuming that our wave functions have decayed once the infinite barriers are reached (see Sec. III A).

A, B, C, and D in Table I are normalization constants. $\alpha = [2m_p^*(V_p - E)]^{1/2}/\hbar$ and $k = (2m_p^*E)^{1/2}/\hbar$. By ensuring the continuity of the amplitudes of the wave functions and the probability current at the boundaries, the following transcendental equations for the energy *E* are obtained.

For even solutions,

$$\tan\left(\frac{(2m_p^*E)^{1/2}a}{2\hbar}\right) = \left(\frac{V_p - E}{E}\right)^{1/2}.$$
 (A1)

For odd solutions,

$$\cot\left(\frac{(2m_p^*E)^{1/2}a}{2\hbar}\right) = -\left(\frac{V_p - E}{E}\right)^{1/2}.$$
 (A2)

These equations were solved by using the numerical root finding algorithm provided with the NAG package. Once the energies are obtained, α and k follow from simple substitution. The normalization constants follow from the usual normalization procedures once all of the other parameters are known.



FIG. 20. Schematic diagram of the potential used to generate the unbound basis states used in the calculations.

| Region | Even solutions | Odd solutions |
|----------------------------------|---------------------------------|--------------------------------|
| -L/2 < x < -a/2 $-a/2 < x < a/2$ | $-A \sin[k'(x-L)]$ B cos(kx) | $C \sin[k'(x-L)]$ D cos(kx) |
| $a/2 \le x \le L/2$ | $A\sin[k'(x-L)]$ | $-C\sin[k'(x-L)]$ |

TABLE II. Table of the forms of the unbound state solutions for the finite square-well potential.

2. Unbound states

The problem for unbound states is somewhat less straightforward. We assume that the finite well (of width 2a) is embedded within an infinite well (of width 2L), and that we may set the wave function outside the infinite well to zero (see Fig. 20). Then there are again three regions which have different forms of solutions, which take the forms given in Table II.

The wave vector inside the dots is given by $k = (2m^*E)^{1/2}/\hbar$, and $k' = [2m^*(E-V_p)]^{1/2}/\hbar$. The solution is obtained by again invoking the continuity of the wave

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function amplitude and the probability current at the boundaries, and the following transcendental equations are found for the even and odd solutions, respectively,

$$\left(\frac{E-V_p}{E}\right)^{1/2} = -\tan\left(\frac{a(2m^*E)^{1/2}}{\hbar}\right) \tan\left(\frac{(a-L)[2m^*(E-V_p)]^{1/2}}{\hbar}\right),$$
(A3)
$$\left(\frac{E-V_p}{E}\right)^{1/2} = \cot\left(\frac{a(2m^*E)^{1/2}}{\hbar}\right) \tan\left(\frac{(a-L)[2m^*(E-V_p)]^{1/2}}{\hbar}\right).$$
(A4)

Equations (A3) and (A4) are solved for E, and then the wave vectors and normalization constants follow on as before.

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