PCCP

PAPER

Cite this: Phys. Chem. Chem. Phys., 2014, 16, 10669

Received 21st January 2014, Accepted 14th April 2014

DOI: 10.1039/c4cp01298d

www.rsc.org/pccp

1. Introduction

Nanocrystal sensitized solar cells (NCSCs) have been under extensive investigation due to their potentially high photoconversion efficiencies and reduced cost.¹⁻⁸ Various strategies have been employed to maximize NCSC device efficiencies, however, the current record efficiencies remain around \sim 5.4%, which is significantly lower than the predicted limit of ~20%.^{9,10} To create NCSCs, two prominent techniques have arisen: the first relies on the fine tuning of the nanocrystal sensitizer size to control the band-gap energy using colloidal chemistry techniques,^{11–17} the second is via successive ionic layer adsorption and reaction (SILAR).¹⁸⁻²³ Within these two approaches, emphasis has been placed upon the nanocrystal-metal oxide (NC-MO) heterojunction in order to better understand the physical underpinnings of the charge transfer process.^{2-3,13,15,19,24}

In this report we examine the electron transfer process between SILAR CdS-TiO₂ heterojunctions, and propose a means of enhancing ultrafast charge transfer using laser processing to modify the morphology and crystalline phase of the MO.

Enhancing photo-induced ultrafast charge transfer across heterojunctions of CdS and laser-sintered TiO₂ nanocrystals

Bryan T. Spann, †^a S. Venkataprasad Bhat, †‡^a Qiong Nian, ^b Kelly M. Rickey, ^a Gary J. Cheng,^b Xiulin Ruan*^a and Xianfan Xu*^a

Enhancing the charge transfer process in nanocrystal sensitized solar cells is vital for the improvement of their performance. In this work we show a means of increasing photo-induced ultrafast charge transfer in successive ionic laver adsorption and reaction (SILAR) CdS-TiO₂ nanocrystal heterojunctions using pulsed laser sintering of TiO₂ nanocrystals. The enhanced charge transfer was attributed to both morphological and phase transformations. At sufficiently high laser fluences, volumetrically larger porous networks of the metal oxide were obtained, thus increasing the density of electron accepting states. Laser sintering also resulted in varying degrees of anatase to rutile phase transformation of the TiO₂, producing thermodynamically more favorable conditions for charge transfer by increasing the change in free energy between the CdS donor and TiO₂ acceptor states. Finally, we report aspects of apparent hot electron transfer as a result of the SILAR process which allows CdS to be directly adsorbed to the TiO₂ surface.

> We focus on SILAR heterojunctions because, unlike colloidal synthesis of nanocrystals, it does not require the use of surfaceattached organic ligands. Organic ligands have been found in several reports to act as relaxation and transfer pathways for excitons.^{25–27} The quenching of excitons through organic ligands in NCSCs is detrimental to overall device performance.15 Furthermore, the ultrafast charge transfer between NC and MO determined from the ultrafast measurement is ambiguous when ligands are present, because it is difficult to discern whether electrons are transferring to the desired MO, to the various organic ligands attached to the surface, or relaxing through the other non-radiative phenomenon such as surface trap states or Auger-thermalization.^{26,28-30} On the other hand, when SILAR is used, ultrafast charge transfer can be more easily understood due to the absence of ligands. The presence of organic ligands also creates an additional dielectric barrier of which electrons must overcome in order to transfer to the MO.^{15,17} Tisdale and Zhu pointed out that the thickness of the dielectric barrier would dictate the potential for hot-electron and band-edge charge transfer. Shorter distances between NCs and MO would enhance electronic coupling for hot-electron charge transfer.¹⁷ For SILAR heterojunctions the resistive dielectric layer is not present, thus providing more favorable interfacial conditions for the hot-electron transfer process. Various research groups have shown evidence for hot-electron transfer, e.g., Tisdale and Zhu reported hot-electron transfer in monolayer PbSe QDs coupled to TiO₂,¹⁶ Sambur *et al.* and more recently Yang *et al.* have reported electron transfer occurring within 50 fs³¹ and 6.4 fs³² between

CHEMISTRY

^a Birck Nanotechnology Center and School of Mechanical Engineering,

Purdue University, 1205 West State St., West Lafayette, Indiana 47907, USA. E-mail: xxu@purdue.edu, ruan@purdue.edu

^b School of Industrial Engineering, Purdue University, 315 N. Grant St., West Lafayette, Indiana 47907, USA

[†] These authors contributed equally to this work.

[‡] Current address: SRM Research Institute, SRM University, Kattankulathur, Chennai 603203, India

due to the lack of the ligand barrier. A drawback to NC sensitized solar cells is reduced electron mobility once the exciton has disassociated into individual electrons and holes. This is a result of constricted electron transfer pathways between neighboring MO NCs. While investigations into enhanced charge transfer by tuning, e.g., the NC size and shape,^{12,13} and metal oxide chemical composition,^{15,19} fewer works have been done to observe the effects of morphological dependence of the MO,35 and none have studied the effects of the transformation of the MO crystalline phase on ultrafast charge transfer. In order to facilitate enhanced electron mobility, we employ pulsed laser sintering to induce a mesoporous to macroporous structural transformation in the MO NCs. Accompanied by the morphological transformation is an anatase to rutile phase transformation which enhances electron transfer from SILAR adsorbed CdS NCs to the TiO2 MO. In addition to enhanced electron transfer rates, we also show evidence suggesting hotelectron transfer between CdS and TiO2, as a result of this phase/ morphology transformation as well as the close proximity of NC and MO due to the lack of organic ligands.

hot-electrons as a result of the closer proximity of NC and MO

2. Experimental details

2.1. Sample fabrication

To prepare the CdS-TiO₂ SILAR samples, Degussa P25 mixed phase (80% anatase and 20% rutile) TiO2 NCs were used for the MO electron acceptor. The TiO₂ NCs were roughly spherical with diameters on the order of 10-20 nm. The TiO₂ films were deposited on a glass substrate using the doctor blade technique¹⁵ and were found to be approximately 3.5 µm thick (measured via cross-sectional SEM). All TiO₂ films were calcined at 450 °C for one hour. A pulsed KrF (248 nm wavelength) excimer laser with a repetition rate of 10 Hz and pulse duration of 25 ns was used to sinter the TiO2 films with fluences ranging from 50 to 200 mJ cm⁻². The sintering was performed in a 20 mTorr vacuum environment to minimize sintering induced surface defects in the TiO2. The TiO2 NC films were each exposed to the excimer laser for 1 second. After laser irradiation, CdS was adsorbed on the TiO₂ films using the SILAR technique, which involves dipping the TiO₂ films into 0.1 M Cd(ClO₄)₂, followed by a dipping into 0.1 M Na₂S aqueous solutions at room temperature in an inert environment.²¹ Following this, the samples were rinsed with distilled water. For all CdS-TiO₂ samples we repeated this dip-dip-rinse cycle 8 times to ensure full coverage and adequate growth of the CdS nanocrystals.

2.2. Structural and surface characterization

The TiO₂ films were imaged *via* a scanning electron microscope (SEM, Hitachi S-4800). The crystalline phase of the TiO₂ was determined by means of a Raman microscope (Horiba LabRAM HR). After the CdS was adsorbed to the surface of the TiO₂, an energy dispersive X-ray spectroscope (EDX, FEI Philips XL-40 SEM) was used to compare the ratio of Cd to Ti for each sample. After all optical spectroscopy was performed, a transmission electron microscope (TEM, Titan 80–300 kV) was used to image the CdS–TiO₂ interface conditions and verify the crystalline phase. TEM samples were prepared by dispersing the powder scratched from the prepared films in ethanol and placing droplets of the suspension on carbon coated copper TEM grids.

2.3. Linear and nonlinear optical characterization

For the linear absorption spectroscopy, a Perkin Elmer Lambda 950 UV/Vis/NIR spectrophotometer was used. The transient absorption spectroscopy (TAS) experiment employs a traditional non-collinear pump-probe scheme consisting of a Spectra Physics femtosecond pulsed laser amplifier that produces 70 fs pulses at a central wavelength of 800 nm and at a repetition rate of 5 kHz. The probe leg is sent into a Quantronix TOPAS optical parametric amplifier (OPA). The probe wavelength from the OPA was set to 490 nm (tuned to the conduction band-edge of CdS NC). The pump beam is sent through a mechanical chopper rotating at a frequency of 500 Hz, then through a second harmonic crystal to generate 400 nm pulses for sample excitation. The pump and probe are then focused on the sample non-collinearly to $1/e^2$ spot diameters of 320 µm and 150 µm for the pump and the probe, respectively. The pump fluence was set to approximately 80 μ J cm⁻² to ensure the absence of multiparticle Auger recombination.¹⁴ We placed a 450 nm long-pass light filter after the sample to eliminate the scattered pump photons from reaching our detector. In order to help eliminate laser noise, we used a balanced photo-detector for probe and reference beam detection. Additionally, two 10 nm full width half max (FWHM) band-pass filters centered at the 490 nm probe wavelength were used for each side of the balanced photodetector. The 10 nm band-pass filters provide a spectrally integrated signal to cover the variance in band edge energy as a result of the size dispersion of the CdS NCs.

3. Results and discussion

3.1. TiO₂ characterization

The effects of pulsed laser sintering are immediately obvious in surface SEM images of the TiO₂ films. A representative TiO₂ film at a 50 μ m scale is shown in Fig. 1A, revealing variable coverage as a result of the doctor blade technique. Fig. 1B–F show the change in morphology as a result of sintering of the TiO₂ NC films, with B and F representing the two extremes, no sintering (0 mJ cm⁻²) and the higher laser fluence used, 200 mJ cm⁻², respectively. Growth in fused TiO₂ features appears roughly to be 100 nm, 1 μ m, 5 μ m, and 10 μ m for the laser fluences of 50, 100, 150, and 200 mJ cm⁻², respectively.

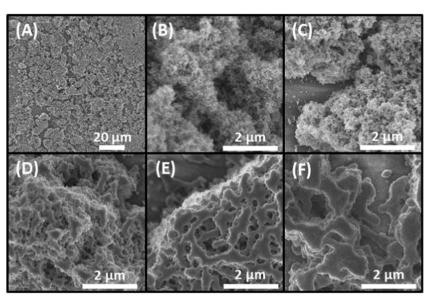


Fig. 1 Surface SEM images of the TiO_2 NC films prior to CdS adsorption. (A) is a representative image of a 50 μ m scaled image of the un-sintered film, (B) is of an un-sintered film, with as prepared TiO_2 nanocrystals (~10 nm diameter), (C), (D), (E), and (F) are films sintered using laser fluences of 50, 100, 150, and 200 mJ cm⁻², respectively.

Furthermore, relatively larger scale networks appear to be forming for sintering cases 100 mJ cm⁻² and above, while the 50 mJ cm⁻² sample shows only highly localized sintered TiO₂ networks. The larger networks of sintered TiO₂ are expected to enhance mobility of electrons once they have transferred into the TiO₂. Furthermore, the density of accepting states is also expected to increase as will be discussed later. All samples, except the 200 mJ cm⁻² sample, show visible porosity in their SEM images.

Raman spectra were collected for the TiO_2 sintered films and are shown in Fig. 2. The anatase (ana) and rutile (rut) phase peaks are labeled with dashed vertical lines. The dominant anatase

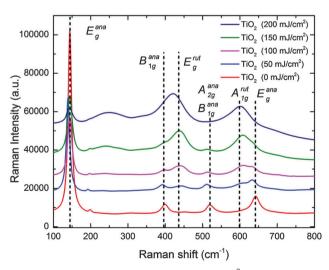


Fig. 2 Raman spectra of un-sintered (0 mJ cm⁻²) and sintered TiO_2 samples using laser fluence of 50, 100, 150, and 200 mJ cm⁻². The dashed vertical lines denote pure phase Raman shift peaks, superscripts (ana) and (rut) represent anatase and rutile phases, respectively.

peaks are assigned to the following modes: $E_{\rm g}^{\rm ana}$ (~150 cm⁻¹ and 640 cm⁻¹), $B_{1g}^{\rm ana}$ (~400 cm⁻¹ and 515 cm⁻¹), and A_{2g}^{ana} (515 cm⁻¹).³⁶ The rutile phase peaks are assigned to the following modes: E_{g}^{rut} (~446 cm⁻¹) and A_{1g}^{rut} (~610 cm⁻¹).³⁷ The Raman data show that as the sintering fluence is increased, the TiO₂ films undergo varying degrees of phase transformation from anatase to rutile. This is evident by viewing the Raman spectra of the un-sintered sample (0 mJ cm^{-2} fluence), which shows predominantly anatase phase Raman modes (also expected from the well-known Degussa P25 TiO₂). As the sintering fluence increases, growth in the E_{g}^{rut} and A_{1g}^{rut} modes and decay in the anatase modes are very clear. Furthermore, once the sintering fluence reaches 150 mJ cm $^{-2}$, the rutile phase begins to dominate the Raman spectra with smaller contributions from the anatase phase peaks. At 200 mJ cm⁻², the Raman spectra display only a low intensity E_{g}^{ana} peak, suggesting almost a full phase transformation. The broad peak centered at 240 cm⁻¹ is attributed to a second-order rutile phonon mode.38

3.2. CdS-TiO₂ adsorption characterization

Using the SILAR technique, described in the sample preparation section below, CdS was adsorbed to the surface of the unsintered and sintered TiO_2 samples. Linear absorption (an approximately 1 cm² sampling area was used to average out the variable coverage of films) and energy-dispersive X-ray spectroscopy (EDX) showed that the relative amount of Cd to Ti was consistent for all samples as shown in Fig. 3A and B, respectively. Fig. 3A also shows the linear absorption data of a CdS–SiO₂ sample that was used in this study as a reference for both linear absorption measurements and more importantly for transient absorption measurements. The offset of the CdS–TiO₂ samples is a result of light scattering from the TiO₂ surface, which increases for longer wavelengths under higher

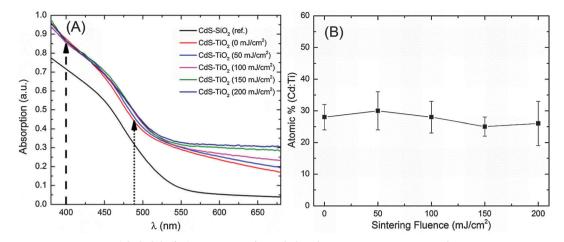


Fig. 3 (A) Linear absorption spectra of CdS–SiO₂ (reference sample) and CdS–TiO₂ samples under various TiO₂ sintering conditions. Also shown are pump (dashed arrow) and probe (dotted arrow) wavelengths for the transient absorption measurements. (B) Relative amounts of Cd to Ti for the samples; ratios were obtained from EDX data.

sintering conditions. The EDX data presented in Fig. 3B show the relative amount of Cd to Ti for each sample. The similar adsorption percentages indicate that sintering does not deteriorate the adsorption capability. It also allows us to directly compare the charge transfer in each sample on an equal basis.

Transmission electron microscopy (TEM) was used to observe the nanoscale interfacial conditions that exist between the CdS and TiO₂. Fig. 4 shows both the low resolution and high resolution images obtained for samples with unsintered and sintered

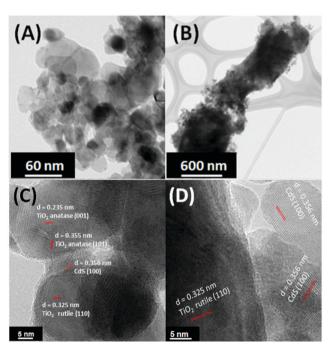


Fig. 4 Low resolution TEM images of CdS adsorbed to TiO₂, (A) before sintering, (B) after sintering showing the change in morphology of TiO₂ particles. High resolution TEM images (C) before sintering, and (D) after sintering with the crystal planes marked for rutile and anatase TiO₂ indicating the phase transformation; the adsorbed CdS are also marked, for clarity and to reveal the interface between CdS and TiO₂.

 TiO_2 particles (150 mJ cm⁻² sintering fluence). The growth of TiO₂ particles upon laser sintering is shown by the relatively larger size of the sintered particles in Fig. 4B when compared with the unsintered particles in Fig. 4A. A high resolution TEM image of the unsintered sample is shown in Fig. 4C, with the crystal planes noted for both the anatase and rutile phases of TiO_2 . We identify the (001) and (101) planes of anatase TiO_2 with interplanar spacings of d = 0.235 and 0.355 nm, respectively. Fig. 4D shows the high resolution TEM image of the 150 mJ cm⁻² sintered case and we are able to identify the (110) plane of rutile phase TiO₂ with an interplanar spacing of d = 0.325 nm. For Fig. 4C and D we mark the (100) CdS planes with d = 0.356 nm. Fig. 4D (the sintered sample) had only rutile phase crystal planes indicating the phase transformation of TiO₂. Furthermore, the interfacial condition between the sintered TiO₂ particles of bigger size also appears to show larger surface area through which electron could transfer from CdS to the TiO₂.

3.3. Transient absorption

Transient absorption spectroscopy (TAS) measurements were performed using a pump wavelength of 400 nm (3.1 eV) and a probe wavelength tuned to the conduction band-edge of the CdS NCs at 490 nm (2.53 eV); see Fig. 3A. The conduction band-edge of the CdS was determined by first probing at 500 nm with 10 nm band-pass filters and all samples' TAS signals exhibited photo-induced absorption features which suggests below band-edge probing. The probe wavelength was then decreased to 490 nm with 10 nm band-pass filters and all samples' TAS signals showed no photo-induced absorption, suggesting that the probe photons are directly probing the average band-edge energy (we have demonstrated this in our earlier studies, see ref. 29 and 30 for more details). By probing at the conduction band-edge we are able to directly monitor the population of this electronic level, therefore, the depopulation of this level corresponds to electron transfer or recombination with the hole. In order to estimate the charge transfer rate from

the TAS signals, we employ a similar strategy to other studies and use a reference sample with SiO_2 in the place of TiO_2 .^{15,19,39} The TAS signal from the CdS– SiO_2 sample will provide purely CdS exciton relaxation; *i.e.*, without charge transfer because of the insulator nature of SiO₂ (9.2 eV band-gap energy).¹⁵

TAS data of CdS-SiO₂ superpose several photo-physical phenomena, namely once the pump photons arrive at the sample, an exciton is generated and because the probe beam is tuned to the conduction band-edge of the CdS, this energy state is already occupied; this leads an increase in transmission. This effect is caused by state-filling induced bleach.40 Another consequence of this is an increase in stimulated emission and spontaneous emission which also gives rise to increased transmission.²⁵ As the probe pulse is delayed with respect to the pump pulse, electrons and holes begin to relax non-radiatively to lower energy surface-trap-states, followed by radiative recombination of electrons and holes.²³ For the samples with CdS adsorbed to TiO₂, the same exciton relaxation mechanisms are active in the CdS. However, an additional electron relaxation pathway arises via transfer to the TiO₂. This allows us to quantitatively analyze the TAS relaxation data by considering the difference in the relaxation rate between the CdS TiO₂ sample and the CdS-SiO₂ reference sample. The electron transfer rate $(k_{\rm ET})$ between CdS and TiO_2 , which may be calculated as,^{19,39}

$$k_{\rm ET} = \frac{1}{(\tau)_{\rm CdS-TiO_2}} - \frac{1}{(\tau)_{\rm CdS-SiO_2}}$$
(1)

where $(\tau)_{CdS-TiO_2}$ and $(\tau)_{CdS-SiO_2}$ are the relaxation times for the CdS-TiO₂ sample and the CdS-SiO₂ reference sample, respectively. The results of the TAS absorption data are presented in Fig. 5, with the inset showing the first 40 ps of the decay. In order to account for the variable coverage in the films (e.g., see Fig. 1A) we used relatively large pump and probe spot sizes of 320 µm and 150 µm, respectively. Furthermore, the TAS results presented in Fig. 5 represent the average of five measurements on each sample. The error-bars included on each line represent the standard deviation between five different spatial locations on the sample. From Fig. 5, the CdS-SiO₂ sample shows the slowest relaxation (due to absence of charge transfer). Furthermore, a background photoluminescence (PL) is present for this sample, shown by the non-zero baseline in the TAS signal. We have eliminated the possibility of this signal being a result of scattered pump and probe photons by using light filtration described in the Experimental details section. The PL background of the CdS-SiO₂ is likely to be present based on relaxation of the electrons into long-lived surface trap states that exist due to deficiently passivated surfaces and the dangling bonds that are present. Additionally, we believe that a photocharging effect could be present which can also lead to long lived PL features.⁴¹ The background PL signal contributes to the entire time signal, therefore, for the analysis of charge transfer, the background signal is subtracted and the signal is re-normalized for the charge transfer calculations.

For the CdS–TiO₂ samples, significant charge transfer is evident for all samples (shown by faster exponential decay in Fig. 5). Furthermore, the PL background has been predominantly

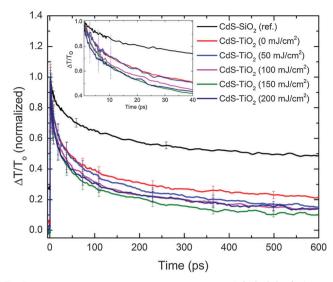


Fig. 5 Transient absorption spectroscopy data of CdS–SiO₂ (reference sample) and CdS–TiO₂ samples under all sintering conditions. Note that the CdS–SiO₂ is an un-sintered sample. The data represent the average of five spatial locations on the sample with the error bars representing one standard deviation between the various spatial locations. The inset shows the first 40 ps.

quenched for all CdS-TiO₂ samples, which is indicative of band-edge charge transfer.¹⁹ Fig. 5 also shows the sintering effects on the TAS; as the sintering laser fluence increases, the charge transfer increases up to the laser fluence of 150 mJ cm⁻². Using the laser fluence of 200 mJ cm⁻², the charge transfer reduces relative to the 150 mJ cm⁻² case and more closely resembles the 100 mJ cm⁻² case.

Biexponential fits of the TAS data were used to quantify the fast and slow components of relaxation. For the CdS–SiO₂ sample the fast component corresponds to relaxation of excitons into surface trap states and the slow component corresponds to radiative recombination, these two processes define the intrinsic decay of the CdS.^{19,39} For the CdS–TiO₂ samples, the same fast and slow processes exist with the addition of the charge transfer pathway to the TiO₂.^{19,39} We have considered both fast and slow processes separately in our analysis, by segregating the components of the biexponential fits of the normalized TAS data:

$$\Delta T/T_0 = C_1 e^{-t/\tau_1} + C_2 e^{-t/\tau_2}$$
(2)

where, C_i are the pre-exponential factors which correspond to the relative amplitude of each lifetime, and τ_i are the decay times associated with each term.^{19,39} We define the first term of the biexponential fit $(C_1 e^{-t/\tau_1})$ to represent the fast component, the second term $(C_2 e^{-t/\tau_2})$ represents the slow component. From eqn (2) we can define two new equations in the form of eqn (1) to calculate the electron transfer rates that occur during the fast and slow timescales separately, those are:

$$k_{\rm F,ET} = (1/\tau_1)_{\rm CdS-TiO_2} - (1/\tau_1)_{\rm CdS-SiO_2}$$

$$k_{\rm S,ET} = (1/\tau_2)_{\rm CdS-TiO_2} - (1/\tau_2)_{\rm CdS-SiO_2}$$
(3)

Paper

Published on 14 April 2014. Downloaded by Purdue University on 15/05/2014 14:57:33.

where the subscripts F,ET and S,ET denote the electron transfer during the fast and slow timescales, respectively. Note: here we are not suggesting that there are fast and slow electron transfer processes, rather we are using the two terms to portray the relative amount of electrons transferring during the two different timescales in this measurement. We have employed this method to better portray the relative contributions of charge transfer over the two timescales. The biexponential fitting parameters and electron transfer rate calculation results are shown in Table 1. We can see from Table 1 that the slow radiative recombination component (*i.e.*, C_2) is the dominant physical event for the CdS-SiO₂ sample, whereas for the CdS-TiO₂ samples, the contribution of the fast component in the decay (*i.e.*, C_1) is more dominant when compared with the intrinsic decay of the CdS-SiO₂ sample. In order to depict the results more clearly, the results of the electron transfer rate $(k_{\rm ET})$ calculations are presented in Fig. 6. Again, in Fig. 6, the laser sintering fluence of 150 mJ cm^{-2} yields the highest charge transfer rate for both fast and slow components of the electron transfer process. In fact, when comparing it to no-sintering, there is approximately a two-fold increase in the rate of transfer for both fast and slow processes. This two-fold increase in electron charge transfer is significant because hole transfer from QDs has been shown to be on the order of ~ 4 ps for InP QDs coupled with a tetramethyl-p-phenylenediamine hole acceptor, making both electron and hole transfer enhancement crucial to the overall device performance.42 We have also calculated an average electron charge transfer rate using the following expression:19,39

$$\langle k_{\rm ET} \rangle = 1/\langle \tau \rangle_{\rm CdS-TiO_2} - 1/\langle \tau \rangle_{\rm CdS-SiO_2}$$
 (4)

where the weighted average relaxation times can be calculated as: 19,39

$$\langle \tau \rangle = \frac{\sum_{i=1}^{n} C_{i} \tau_{i}^{2}}{\sum_{i=1}^{n} C_{i} \tau_{i}} = \frac{C_{1} \tau_{1}^{2} + C_{2} \tau_{2}^{2}}{C_{1} \tau_{1} + C_{2} \tau_{2}}$$
(5)

where the C_i and τ_i come from the biexponential fit function. The weighted average electron transfer rate is often used, ^{19,39,43} but here we show that some insight is lost when only this value is reported. For instance, the results of the $\langle k_{\rm ET} \rangle$ calculation are found in Table 1 and show that the electron transfer rate is dominated by the slower component of the relaxation. However, as evident from Table 1, Fig. 5 and 6, both timescales show variance in the charge transfer rate.

3.4. Insights from Marcus theory

In order to understand the mechanisms leading to increased electron transfer, we must consider the energy band alignments for each of the materials in the charge transfer process. Shown in the inset of Fig. 6 is the vacuum energy level containing the relative energy level differences in the conduction band minimum (CBM) and valence band maximum (VBM) for the constituent materials. As shown in the Raman spectroscopy results (Fig. 2) and also the TEM results (Fig. 4), the relative amount of the rutile phase to anatase phase in the TiO₂ is increased as the sintering fluence is increased. The rutile phase CBM is approximately 0.2 eV lower than the anatase phase CBM,^{37,44} thus an increasing electronic potential between donor and accepting states exists between the CdS and TiO₂ as the sintering increases. This acts as a stronger driving force for electrons to be transferred. Furthermore, it has been shown that by introducing an intermediate level anatase TiO2 structure, once electrons are transferred to the rutile TiO₂ it reduces the potential for recombination.37

The fundamental reasons for the increase in charge transfer can be clarified further using Marcus theory. To elucidate the details of charge transfer in NC–MO systems, various groups have successfully applied the many-state non-adiabatic Marcus model.^{15,17,35,45} Using Marcus theory, the electron transfer rate between donor and acceptor states can be calculated as,¹⁵

$$k_{\rm ET} = \frac{2\pi}{\hbar} \int_{-\infty}^{\infty} \rho(E) |\bar{H}(E)|^2 \frac{1}{\sqrt{4\pi\lambda k_{\rm B}T}} \exp\left(-\frac{(\lambda + \Delta G + E)^2}{4\lambda k_{\rm B}T}\right) dE$$
(6)

where $\rho(E)$ is the energy dependent density of accepting states (DOS) in the MO. $\bar{H}(E)$ is an overlap matrix element, λ is the system reorganizational energy, $k_{\rm B}$ is Boltzmann's constant, \hbar is Planck's constant, T is the temperature, and ΔG is the change in system free energy. Although all terms in eqn (6) affect the charge transfer rate, for this study $\rho(E)$ and ΔG dominate the difference in charge transfer as a result of the altered phase MO. The wavefunction overlap matrix element $\bar{H}(E)$ is typically assumed to be independent of energy and the reorganizational energy λ is expected to be independent of the MO and depend only on the CdS sensitizer.^{15,17,35,45} To establish the sources of the increase in charge transfer, we first consider the DOS.

 Table 1
 Biexponential fitting and charge transfer rate results from transient absorption data

	Sintering fluence (mJ cm^{-2})	C_1	τ_1 (s)	C_2	τ_2 (s)	$k_{\mathrm{F,ET}} \left(\mathrm{s}^{-1} \right)$	$k_{\mathrm{S,ET}} \left(\mathrm{s}^{-1} \right)$	$\langle k_{ m ET} angle ~ \left({ m s}^{-1} ight)$
CdS-SiO2 ^a	N.A.	0.28	2.76×10^{-11}	0.72	8.13×10^{-10}	N.A.	N.A.	N.A.
CdS-TiO ₂	0 50 100 150 200	0.47 0.48 0.52 0.56 0.53	$\begin{array}{c} 1.60 \times 10^{-11} \\ 1.72 \times 10^{-11} \\ 1.48 \times 10^{-11} \\ 1.20 \times 10^{-11} \\ 1.22 \times 10^{-11} \end{array}$	0.53 0.52 0.48 0.44 0.47	$\begin{array}{c} 6.64 \times 10^{-10} \\ 4.65 \times 10^{-10} \\ 4.44 \times 10^{-10} \\ 3.35 \times 10^{-10} \\ 4.47 \times 10^{-10} \end{array}$	$\begin{array}{c} 2.63 \times 10^{10} \\ 2.17 \times 10^{10} \\ 3.11 \times 10^{10} \\ 4.69 \times 10^{10} \\ 4.55 \times 10^{10} \end{array}$	$egin{array}{llllllllllllllllllllllllllllllllllll$	$\begin{array}{l} 2.91 \times 10^{08} \\ 9.77 \times 10^{08} \\ 1.08 \times 10^{09} \\ 1.86 \times 10^{09} \\ 1.06 \times 10^{09} \end{array}$

^a The background PL signal has been subtracted and the signal is re-normalized and fitted for the charge transfer calculations.

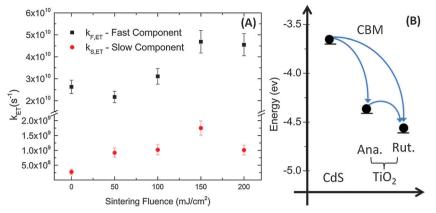


Fig. 6 (A) Fast and slow components of calculated electron transfer rates (k_{ET}) from the adsorbed CdS nanocrystals to TiO₂. (B) Shows charge transfer pathways on a vacuum energy level diagram, *e.g.*, the band-edge electron in the CdS nanocrystal can transfer to anatase (ana) or to rutile (rut) phase TiO₂ or sequentially transfer from CdS to anatase then to rutile TiO₂.

In bulk-like MOs, assuming a parabolic dispersion, $\rho(E)$ can be described by:¹⁵

$$\rho(E) = V_0 \frac{(2m_{\rm e}^*)^{3/2}}{2\pi\hbar^3} \sqrt{E}$$
(7)

where V_0 is the localized electron acceptor volume of the crystal. We expect V_0 to not vary between sintering conditions because the interfacial conditions between the donor and the acceptor are similar. The m_e^* variable is the effective mass of the electrons in TiO₂. Under higher sintering conditions, we expect an increase in $\rho(E)$ for the TiO₂ because of the increase in electron effective mass, $m_e^* \sim 1m_0$ and $m_e^* \sim 8-20m_0$ for anatase and rutile, respectively.⁴⁶ The increased DOS in the MO results higher potential for electron transfer. The other dominant modification, as a result of the phase transformation, is the change in free energy ΔG . Tvrdy *et al.* derived a simple expression to calculate the change in free energy for NC-MO systems,¹⁵

$$\Delta G \approx E_{\rm MO} - E_{\rm 1Se} + (1+C) \frac{e^2}{\varepsilon_{\rm QD} R_{\rm QD}} + \frac{e^2}{2R_{\rm QD}} \left(1 + \frac{C}{\varepsilon_{\rm QD}}\right) - \frac{e^2}{4(R_{\rm OD} + h)} \frac{\varepsilon_{\rm MO} - 1}{\varepsilon_{\rm MO} + 1}$$
(8)

where $E_{\rm MO}$ and $E_{1\rm Se}$ are the electron energies at the CBM of the TiO₂ and CdS, respectively, *e* is the fundamental charge constant, $R_{\rm QD}$ is the average radius of the CdS NCs, *h* is the spatial separation between the donor and the acceptor, $\varepsilon_{\rm QD}$ and $\varepsilon_{\rm MO}$ are the dielectric permittivity of the CdS and TiO₂, respectively, and finally $C \approx 0.786$ is a numerical integration constant.¹⁵ As mentioned previously, the anatase to rutile phase conversion results in a 0.2 eV reduction in $E_{\rm MO}$, resulting in CBM values of -4.41 eV and -4.61 eV, respectively.^{37,44} The dielectric permittivity values of the TiO₂ are $\varepsilon_{\rm MO} \approx 9.9^{15}$ and 6.33^{46} for anatase and rutile, respectively. Both changes result in modifications to ΔG .

Taking only the energy dependent parameters and considering the cases of pure anatase and rutile phases, we can use the experimentally obtained parameters $R_{\rm QD}$ (~7 nm) and $E_{\rm 1Se}$ (~-3.7 eV)⁴⁷ and values from the literature discussed above to calculate a relative increase in band-edge charge transfer as a result of the difference in ΔG and effective mass, *i.e.*,

$$\frac{k_{\rm ET,rut}}{k_{\rm ET,ana}} \propto \left(\frac{m_{\rm e,rut}^*}{m_{\rm e,ana}^*}\right)^{3/2} \frac{\int_{-\infty}^{\infty} \sqrt{E} \exp\left(-\frac{\left(\lambda + \Delta G_{\rm rut} + E\right)^2}{4\lambda k_{\rm B}T}\right) dE}{\int_{-\infty}^{\infty} \sqrt{E} \exp\left(-\frac{\left(\lambda + \Delta G_{\rm ana} + E\right)^2}{4\lambda k_{\rm B}T}\right) dE}$$
(9)

Using eqn (8), we find that $\Delta G_{\text{ana}} = -0.546 \text{ eV}$ and $\Delta G_{\text{rut}} = -0.748 \text{ eV}$. Using these values for eqn (9) with $\lambda = 10 \text{ meV}$,¹⁷ we find that $\frac{k_{\text{ET,rut}}}{k_{\text{ET,ana}}} \approx 26.59$ to 105.1, indicating a large increase in the theoretical charge transfer rate as a result of the change in free energy and the change in effective mass between the two phases. The effective mass contributions provide the large variance in the ratio as a result of $m_e^* \sim 8-20m_0$ for the rutile TiO₂.

The eventual reversal of charge transfer increase when the sintering laser fluence surpasses 150 mJ cm⁻² could be explained by an unfavorable decrease in TiO₂ adsorption surface area. For the 200 mJ cm⁻² fluence, the microporous structure appears to be lost (see Fig. 1F), suggesting that the CdS would have less surface area to adsorb to. Furthermore, because the EDX data show very similar Cd:Ti quantity levels, the CdS could be adsorbed to other CdS and not to the TiO₂ (and this appears to be the case according to Fig. 4B, showing the smaller CdS NCs stacked on one another), this would result in a reduction in charge transfer.

3.5. Aspects of hot electron transfer

In addition to the band-edge charge transfer, we also analyze the TAS signal rise times. The rise time of the TAS signals corresponds to the hot-exciton relaxation time in the CdS NCs. The TAS data are presented in Fig. 7A with the corresponding rise time values presented in Fig. 7B. Note that for Fig. 7A the

PCCP

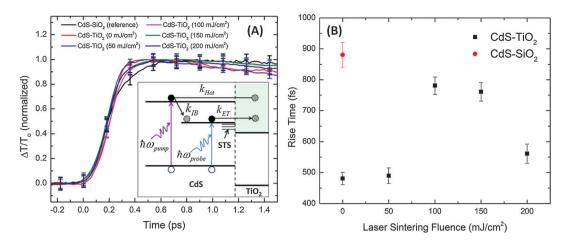


Fig. 7 Normalized early rise time transient absorption signals with background photoluminescence subtracted from the CdS–SiO₂ sample for direct comparison, and corresponding rise time values as a function of sintering fluences shown in (A) and (B), respectively. Inset of (A) shows potential early-time relaxation pathways, including hot electron transfer (k_{hot}), intraband relaxation rate (k_{IB}), and cold/band-edge electron transfer (k_{ET}).

background long-lived surface trap photoluminescence (STPL) has been subtracted out from the CdS-SiO₂ signal to provide a direct comparison between rise times. This signal is likely a result of long-lived deep trap emission or potentially photocharging resulting from prolonged UV exposure as mentioned previously, however this subtraction will not modify the rise time analysis. From Fig. 7B it is evident that there is a considerable difference in rise times when comparing the various sintering fluences. It is evident that the 0 and 50 mJ cm^{-2} samples have similar rise times of approximately 480 fs, while the 100 and 150 mJ cm⁻² cases show an increase in rise time to roughly 750 to 780 fs. The increase in TAS rise time could potentially be explained as a lack of hot-electron transfer from the CdS to TiO₂. To analyze this issue, consider the inset of Fig. 7A, the rise time of the signal denotes the time for electrons high in the conduction band of the CdS to relax down to the band-edge, this is the intraband relaxation rate $(k_{\rm IB})$. If all of the electrons are allowed to relax to the CBM of the CdS, this would correspond to the longest possible saturation time, *i.e.* the intrinsic rise time of the CdS. As we can see from Fig. 7B, the intrinsic rise time of the CdS-SiO₂ is nearly 900 fs, which is the longest rise time of all of the samples tested. Furthermore, as mentioned above, we employed a low pump fluence of 80 μ J cm⁻², therefore, we can neglect the Auger-recombination.¹⁴ This suggests that a decrease in rise time corresponds to transferring of a small portion of the hot-electron population prior to reaching the band-edge (shown by the $k_{\rm hot}$ pathway in the inset of Fig. 7A), thus causing the band-edge to saturate faster. Based on this argument, the unsintered and the 50 mJ cm^{-2} samples show the most hot-electron transfer. For this to be true a key assumption is that the same number of CdS NC must be excited for all cases. Because the linear absorption spectra of the samples all exhibit nearly identical trends and the EDX data show similar amounts of Cd to Ti ratios for all samples, it is likely that this assumption is valid. An important observation pertaining to this assumption is that because the linear absorption and Cd: Ti ratios remain constant this requires buildup of

CdS on CdS under higher sintering conditions as a result of the reduction in TiO₂ adsorption area. The build up of CdS is evident in Fig. 4B, showing CdS not directly adsorbed to the TiO₂ crystallite. Based on this argument, it is fair to assume that if some of the electron population are transferred to the TiO₂ prior to thermalization, this would show up as a reduction in overall population in the TAS measurement, and thus a reduction in the overall TAS signal from the band-edge. Based on our measurements, the signal reduction is within the errorbars of the experiment. Therefore, we can conclude that if there is hot-electron transfer occurring, it only contributes almost negligibly to the overall TAS signal. Considering this observation, we would also like to point out that changes in rise time of the TAS signal could also be the result of inter-particle coupling between the CdS NCs which is expected to change based on the TiO₂ morphology. Inter-particle coupling between CdSe NCs has been demonstrated to have an effect on the intraband relaxation, see for example, the work reported by Yang et al.48 and Gao et al.49

We believe that the apparent hot-electron transfer can be explained via the synthesis process of SILAR. Tisdale and Zhu showed that by considering the radial probability density distribution for a NC, the 2S electron probability density distribution extends substantially farther outside the NC surface than the lower energy 1S probability density distribution.¹⁷ Therefore, hot-electrons have longer range interaction when compared to thermalized electrons. SILAR, as mentioned above, is a direct adsorption technique that is absent of organic ligands, which allows the CdS to be directly attached to the TiO₂ (see TEM images in Fig. 4). Common organic ligands and molecular linkers used such as dodecylamine and 2-mercaptopropionic acid (2-MPA) have linker lengths on the order of 0.53 nm.³⁵ The radial probability distribution is exponentially reduced at distances greater than approximately 0.2 nm outside the donor NC, thus the direct electronic interaction between the donor and the acceptor for the case of an addition dielectric layer is also drastically reduced.¹⁷ This could explain why we observe

hot-electron transfer while Tvrdy *et al.*¹⁵ did not detect such a phenomenon. Furthermore, we believe that this proximity argument can be applied to explain why we observe differences in rise times between our samples. For example, by sintering the TiO_2 , we effectively reduce the adsorption surface area for the SILAR process, this means that the unsintered sample has the most adsorption surface area, thus more CdS can be directly attached to TiO_2 , whereas, the sintered samples have less adsorption area and the CdS can adsorb on other CdS which would eliminate the possibility of hot electron transfer for the spatially separated CdS.

4. Conclusion

To conclude, we used pulsed laser sintering of TiO₂ nanocrystals to enhance charge transfer between CdS and TiO₂. The enhancement in charge transfer is found to be due to the phase transformation induced by the sintering process which increases the relative amount of rutile phase to anatase phase in the TiO₂. This phase transformation increases the electronic potential between donor and accepting states in a thermodynamically favorable way, by increasing the TiO₂ DOS ($\rho(E)$) and reducing the change in free energy (ΔG), thus increasing the charge transfer from the CdS to TiO₂. In addition, we found potential hot-electron transfer for the unsintered sample and the 50 mJ cm⁻² laser sintered sample which was inferred based on TAS signal rise time arguments. The hot-electron transfer is believed to be a result of the direct adsorption of the CdS to the TiO₂ which allows for better electronic interaction between the two materials. For the band-edge electron transfer, it is shown that the 150 mJ cm^{-2} laser sintered sample provided the best overall performance in terms of charge transfer. Our findings suggest that optimal sintering conditions could be used to maximize both hot-electron transfer and bandedge electron transfer from CdS to TiO₂.

Acknowledgements

Support for this work was provided by the National Science Foundation and is gratefully acknowledged. The authors wish to thank Dr Sergey Suslov for assistance with TEM imaging, James Mitchell for assistance with SEM imaging, and Yan Wang for discussions on the charge transfer calculations.

Notes and references

- 1 M. Grätzel, Nature, 2001, 414, 338-344.
- 2 P. Kamat, J. Phys. Chem. Lett., 2013, 4, 908-918.
- 3 P. V. Kamat, J. Phys. Chem. C, 2007, 111, 2834–2860.
- 4 A. Nozik, *Physica E*, 2002, **14**, 115–120.
- 5 H. Zhu and T. Lian, Energy Environ. Sci., 2012, 5, 9406–9418.
- 6 C. A. Nelson, N. R. Monahan and X.-Y. Zhu, *Energy Environ. Sci.*, 2013, **6**, 3508.
- 7 M. C. Beard, J. Phys. Chem. Lett., 2008, 2, 1282-1288.
- 8 G. Conibeer, I. Perez-Wurfl, X. Hao, D. Di and D. Lin, Nanoscale Res. Lett., 2012, 7, 193–199.

- 9 P. K. Santra and P. V. Kamat, J. Am. Chem. Soc., 2012, 134, 2508–2511.
- 10 S. Rühle, M. Shalom and A. Zaban, *ChemPhysChem*, 2010, 11, 2290–2304.
- 11 K. Zídek, K. Zheng, M. Abdellah, N. Lenngren, P. Chábera and T. Pullerits, *Nano Lett.*, 2012, **12**, 6393–6399.
- 12 I. Robel, M. Kuno and P. V. Kamat, J. Am. Chem. Soc., 2007, 129, 4136–4137.
- 13 D. R. Pernik and K. Tvrdy, J. Phys. Chem. C, 2011, 115, 13511-13519.
- 14 I. Robel, V. Subramanian, M. Kuno and P. V. Kamat, J. Am. Chem. Soc., 2006, **128**, 2385–2393.
- 15 K. Tvrdy, P. A. Frantsuzov and P. V. Kamat, Proc. Natl. Acad. Sci. U. S. A., 2011, 108, 29–34.
- 16 W. A. Tisdale, K. J. Williams, B. a. Timp, D. J. Norris,
 E. S. Aydil and X.-Y. Zhu, *Science*, 2010, 328, 1543–1547.
- 17 W. A. Tisdale and X.-Y. Zhu, Proc. Natl. Acad. Sci. U. S. A., 2011, 108, 965–970.
- 18 C.-H. Chang and Y.-L. Lee, *Appl. Phys. Lett.*, 2007, **91**, 053503–053506.
- 19 J. L. Blackburn, D. C. Selmarten and A. J. Nozik, J. Phys. Chem. B, 2003, 107, 14154–14157.
- 20 T. Zewdu, J. N. Clifford, J. P. Hernández and E. Palomares, Energy Environ. Sci., 2011, 4, 4633-4638.
- 21 Y. Tachibana, K. Umekita, Y. Otsuka and K. Susumu, *J. Phys. Chem. C*, 2009, **113**, 6852–6858.
- 22 H. Lee, H. C. Leventis, S.-J. Moon, P. Chen, S. Ito, S. A. Haque, T. Torres, F. Nüesch, T. Geiger, S. M. Zakeeruddin, M. Grätzel and M. K. Nazeeruddin, *Adv. Funct. Mater.*, 2009, **19**, 2735–2742.
- 23 D. R. Baker and P. V. Kamat, Adv. Funct. Mater., 2009, 19, 805–811.
- 24 N. Guijarro and T. Lana-Villarreal, J. Phys. Chem. C, 2010, 21928–21937.
- 25 P. Kambhampati, Acc. Chem. Res., 2011, 44, 1-13.
- 26 P. Kambhampati, J. Phys. Chem. C, 2011, 115, 22089-22109.
- M. Achermann, M. A. Petruska, S. A. Crooker and
 V. I. Klimov, *J. Phys. Chem. B*, 2003, 107, 13782–13787.
- 28 V. I. Klimov, J. Phys. Chem. B, 2000, 104, 6112-6123.
- 29 B. T. Spann, L. Chen, X. Ruan and X. Xu, *Opt. Express*, 2013, 21, A15–A22.
- 30 B. T. Spann and X. Xu, J. Phys. Chem. C, 2014, 118, 2844-2850.
- 31 J. Sambur, T. Novet and B. Parkinson, *Science*, 2010, 330, 63–66.
- 32 Y. Yang, W. Rodríguez-Córdoba, X. Xiang and T. Lian, *Nano Lett.*, 2012, **12**, 303–309.
- J. Pijpers, R. Koole, W. H. Evers, A. J. Houtepen, S. Boehme, C. de M. Donega, D. Vanmaekelbergh and M. Bonn, *J. Phys. Chem. C*, 2010, **114**, 18866–18873.
- 34 E. Cánovas, P. Moll, S. A. Jensen, Y. Gao, A. J. Houtepen, L. D. A. Siebbeles, S. Kinge and M. Bonn, *Nano Lett.*, 2011, 11, 5234–5239.
- 35 K. Zheng, K. Žídek, M. Abdellah, P. Chábera, M. S. Abd El-sadek and T. Pullerits, *Appl. Phys. Lett.*, 2013, **102**, 163119.
- 36 D. A. H. Hanaor and C. C. Sorrell, J. Mater. Sci., 2010, 46, 855-874.

- 37 J. Li, M. W. G. Hoffmann, H. Shen, C. Fabrega, J. D. Prades, T. Andreu, F. Hernandez-Ramirez and S. Mathur, *J. Mater. Chem.*, 2012, 22, 20472–20476.
- 38 S. P. S. Porto, P. A. Fleury and T. C. Damen, *Phys. Rev.*, 1967, 154, 522–526.
- 39 K. Tvrdy and P. V. Kamat, J. Phys. Chem. A, 2009, 113, 3765-3772.
- 40 V. I. Klimov, Annu. Rev. Phys. Chem., 2007, 58, 635-673.
- 41 L. A. Padilha, J. T. Stewart, R. L. Sandberg, W. K. Bae, W.-K. Koh, J. M. Pietryga and V. I. Klimov, *Acc. Chem. Res.*, 2013, 46, 1261–1269.
- 42 J. L. Blackburn, D. C. Selmarten, R. J. Ellingson, M. Jones, O. Micic and A. J. Nozik, *J. Phys. Chem. B*, 2005, 109, 2625–2631.

- 43 I. Robel, B. A. Bunker and P. V. Kamat, *Adv. Mater.*, 2005, **17**, 2458–2463.
- 44 D. Tsukamoto, Y. Shiraishi, Y. Sugano, S. Ichikawa, S. Tanaka and T. Hirai, *J. Am. Chem. Soc.*, 2012, **134**, 6309–6315.
- 45 K. Žídek, K. Zheng, C. S. Ponseca, M. E. Messing, L. R. Wallenberg, P. Chábera, M. Abdellah, V. Sundström and T. Pullerits, *J. Am. Chem. Soc.*, 2012, **134**, 12110–12117.
- 46 R. Asahi, Y. Taga, W. Mannstadt and A. J. Freeman, *Phys. Rev. B: Condens. Matter Mater. Phys.*, 2000, 61, 7459–7465.
- 47 C. G. Van de Walle and J. Neugebauer, *Nature*, 2003, **423**, 626–628.
- 48 Y. Yang, Z. Liu and T. Lian, Nano Lett., 2013, 13, 3678-3683.
- 49 Y. Gao, E. Talgorn, M. Aerts, M. T. Trinh, J. M. Schins, A. J. Houtepen and L. D. A. Siebbeles, *Nano Lett.*, 2011, 11, 5471–5476.