Single and Multi-Exciton Dynamics in Nanoscale Semiconductors

by

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Abstract

Recent advancements in the synthesis and fabrication of nanoscale semiconductors and optoelectronic devices are revolutionizing an array of industries. Nonlinear spectroscopy is a powerful tool for studying the unique optical and electronic properties of these semiconductors. This thesis describes experiments using a unique multi-dimensional spectrometer to study three semiconductors: double-walled J-aggregate nanotubes, cuprous oxide, and monolayer transition metal dichalcogenides.

First, we use two-dimensional correlation spectroscopy and cryogenic tunneling electron microscopy to correlate the excitonic and structural properties of J-aggregate nanotubes in solution. We observe weak coupling between inner and outer wall excitons in both isolated and bundled nanotubes. We also use two-dimensional rephasing spectroscopy to measure the homogeneous and inhomogeneous linewidths in solution at 295 K and in a glass at 10 K and determine dynamic and static disorder dominate, respectively. In addition, we observe photo-induced damage and recovery of nanotubes in the solid state.

Quantum process tomography is used to unambiguously elucidate the single-exciton dynamics of J-aggregate nanotubes in solution. Inversion of spectroscopic signals from eight transient grating experiments completely characterizes the time evolution of the singleexciton density matrix by determining the process matrix. We confirm the weak coupling of inner and outer wall excitons and observe no contributions from non-secular processes.

Second, we use two-quantum spectroscopy to make the first experimental observation of two-exciton correlations (i.e. biexcitons) in cuprous oxide. The direct measurement of twoexciton correlations supports the proposed mechanism of biexciton-Auger recombination for the efficient suppression of the Bose-Einstein condensation of excitons.

Third, we use two-dimensional rephasing spectroscopy to observe substantial inhomogeneous broadening due to large static disorder in monolayer transition metal dichalcogenides. We also use two-dimensional correlation spectroscopy to reveal interactions of excitons with in-plane optical and acoustic phonons. Lastly, we use one- and two-quantum spectroscopy to measure unprecedentedly large inorganic, biexciton binding energies due to reduced dielectric screening in the atomically thin limit.

The exciton and phonon dynamics revealed in these experiments contribute to our understanding of the elementary excitations in organic and inorganic nanostructured semiconductors. It is our hope these insights will help guide the design of next generation devices utilizing nanoscale semiconductors.

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Chapter 1

Introduction

In recent decades, advancements in the fabrication and synthesis of nanoscale materials has lead to new and interesting physics. The unique properties of nanoscale semiconductors make them promising media for more efficient optical and electronic devices [1, 2]. Indeed, engineers are already utilizing organic and inorganic, nanostructured semiconductors in a variety of devices, including photovoltaics [3, 4, 5], photodetectors [6, 7, 8], light emitting diodes [9, 10, 11, 12, 13], diode lasers [14, 15, 16, 17], optical switches [18, 19, 20], and transistors [21, 22, 23, 24].

In particular, we are interested in new nanostructured materials for application in solar cells and other photovoltaic devices in order to improve light harvesting efficiencies and decrease the economic costs of renewable energy. Solar energy is the most promising source of renewable power as the Earth receives as much energy from the sun in one hour as is consumed globally in a year [25]. Biology has evolved over millions of years to take advantage of this nearly infinite energy source and efficiently harvest light. Much can be learned from photosynthetic systems to help guide the design of better solar cells, photovoltaic devices, and their active media [26, 27, 28, 29]. Most interestingly, experimental evidence suggests the high efficiencies in biological light harvesting systems is due to electronic coherences and the correlated protein dynamics that protect them[30, 31, 32, 33]. Photosynthetic complexes

are rich with both classical and quantum mechanics, which must be carefully tailored in synthetic devices to produce high light harvesting efficiencies [34, 35].

The complexity of biological light harvesting systems often makes them difficult to study. So, there has been great interest in understanding synthetic analogs of photosynthetic complexes, which are much easier to study due to less convoluted dynamics. Multichromophoric systems, such as J-aggregates, have been of particular interest because their electronic couplings and supramolecular structure are similar to naturally occurring chlorosomes [36, 37, 38]. In chapters 5 and 6, we discuss experiments using multi-dimensional optical spectroscopy and transient grating measurements to interrogate electronic coupling and dynamics in a double-walled J-aggregate nanotube.

Multi-dimensional optical spectroscopy is a power tool for measuring electronic coherence and dynamics, electronic interactions with vibrational modes and the environment, homogeneous and inhomogeneous linewidths, multi-particle states, and much more [39, 40, 41]. Multi-dimensional spectroscopy has been used to characterize electronic dynamics in a variety of system including: photosynthetic complexes [42, 43, 44, 45, 46], J-aggregates [47, 48, 49, 50, 51], conjugated polymers [52, 53, 54], singlet fission materials [55], nitrogenvacancy centers [56], monolayer transition metal dichalcogenides [57], inorganic quantum wells [58, 59, 60, 61, 62, 63], quantum dots [64, 65, 66, 67, 68], and much more. However, it is often difficult to separate electronic and vibronic contributions to multi-dimensional signals, even when they are deconvolved over a second frequency axis. Thus, theoretical studies have been undertaken to learn how to accurately interpret spectroscopic signals [69, 70, 71, 72]. Currently, the experimental evidence for electronic coherences in photosynthetic complexes using ultrafast spectroscopy has sparked a debate over whether electronic coherences can occur under incoherent excitation. Both experimental and theoretical studies have provided contradicting results, so the debate continues to be hotly contested [73, 74, 75].

The fundamental electronic excitations in most biological light harvesting systems, Jaggregates, and other semiconductors are correlated electron-hole pairs, called excitons. Excitons are bosonic quasi-particles that can undergo, among other processes, Bose-Einstein condensation [76, 77, 78, 79]. Insight into Bose-Einstein condensation is important for understanding the related processes of superconductivity and superfluidity and their applications in new and more efficient optical and electronic devices [80, 81]. In chapter 7, we discuss the first measurement of biexcitons in cuprous oxide, which leads to the efficient suppression of Bose-Einstein condensation of excitons.

The recent isolation of monolayer transition metal dichalcogenides has brought renewed interest to these now two-dimensional materials [82]. Of particular interest to the engineering community is their transition to a direct band gap semiconductor in the monolayer limit and the emergence of coupled spin and valley physics [83, 84]. Thus, monolayer transition metal dichalcogenides are being employed as active media in a variety of electronic and optical devices [85]. Unprecedented Coulombic interactions in these atomically thin, inorganic semiconductors has also lead to important questions about their applicability in photovoltaic devices as electrons and holes must be separated in order to produce a current [86]. In chapter 8, we discuss experiments to study exciton-exciton and exciton-phonon interactions in monolayer transition metal dichalcogenides.

The outline of the thesis is as follows. In chapter 2, we first discuss the physics of excitons starting with the electronic states of the hydrogen atom and working our way to enhanced electron-hole correlations in nanostructured semiconductors. Then in chapter 3, we discuss the theory and techniques of ultrafast spectroscopy, including linear absorption, pumpprobe, transient grating, and multi-dimensional measurements. In chapter 4, we discuss the specific technical design, capabilities, and limitations of our two-dimensional spectrometer that utilizes diffraction-based spatiotemporal pulse shaping and the calibrations necessary to perform an ultrafast experiment. Chapters 5 through 8 describe experiments exploring exciton-phonon and exciton-exciton interactions in a variety of organic and inorganic, nanostructured semiconductors. In chapter 5, we determine the coupling between inner and outer wall excitons in isolated and bundled double-walled J-aggregate nanotubes in solution. We also explore the processes leading to exciton dephasing in solution at room temperature and in a glass at cryogenic temperature, as well as the photo-induced damage and recovery observed in the solid state. In chapter 6, we utilize quantum process tomography to unambiguously elucidate the single-exciton dynamics in isolated J-aggregate nanotubes. In chapter 7, we make the first direct experimental measurement of biexcitons in cuprous oxide and provide evidence to support a biexciton-Auger mechanism that effectively suppresses the Bose-Einstein condensation of excitons. In chapter 8, we discuss preliminary experiments exploring the strength of exciton-phonon and exciton-exciton interactions in monolayer transition metal dichalcogenides. Finally in chapter 9, we conclude and discuss outlooks for new ultrafast spectroscopic experiments to further our understanding excitons on the nanoscale.

Chapter 2

Excitons in Nanoscale Semiconductors

In chapter 2, we will discuss the physics of the elementary solid state excitation relevant in most optoelectronic devices: the electron-hole quasiparticle called an exciton. In particular, we are interested in exciton properties and dynamics when semiconductors are structured on the nanoscale. Thus, we will work our way from a discussion of the electronic structure of the hydrogen atom, up to the H_2^+ molecule, and finally to semiconductors in the solid state. In doing so, we will describe how excitons are formed and their classifications. Subsequently, we will discuss how excitons interact with each other, other charged species, and crystal lattice vibrations (i.e. phonons). Finally, we will consider how excitonic states are perturbed due to confinement on the nanoscale, starting with a discussion of particle-in-a-box states.

2.1 The Hydrogen Atom

In atoms, electrons occupy localized orbitals with discrete energy levels [87]. The Hamiltonian of the hydrogen atom consists of an electron kinetic energy term and a Coulombic potential energy term describing the electron and proton interaction:

$$\hat{H} = -\frac{\hbar^2}{2\mu}\nabla^2 - \frac{e^2}{4\pi\varepsilon_0 r}$$
(2.1)



Figure 2.1: Hydrogen atomic orbitals. (a) Hydrogen atom diagram, where p^+ and e^- are the proton and electron, respectively, and r is the distance between them. (b) Atomic orbital energies as a function of the principal quantum number n. (c) Spatial dependence of the n = 1 atomic orbital energies on the electron radius r.

where \hbar is Planck's constant, $\mu = m_p m_e / (m_p + m_e) \approx m_e$ is the reduced mass of the proton m_p and electron m_e , ∇^2 is the del operator, e is the charge of an electron, ε_0 is the permittivity of free space, and r is the distance between the electron and proton that is illustrated in figure 2.1a. Solving the Schrödinger equation, $\hat{H}\Psi = E\Psi$, results in a three dimensional wavefunction that is separable into a radial component and spherical harmonics using polar coordinates:

$$\Psi(r,\theta,\phi) = \psi_l^m(r) Y_l^m(\theta,\phi)$$
(2.2)

and is defined by the principal, azimuthal, and magnetic quantum numbers n, l, and m, respectively. The energies of the atomic orbitals or Rydberg states are dependent on the principal quantum number:

$$E_n = -\frac{\hbar^2}{2m_e a_0^2} \frac{1}{n^2} = -\frac{R_y}{n^2}$$
(2.3)

where $a_0 = 4\pi\varepsilon_0\hbar^2/m_e e^2$ is the Bohr radius, which defines the most probable distance between the proton and electron in the lowest lying 1s atomic orbital, and $R_y = \hbar^2/2m_e a_0^2$ is the Rydberg constant. Figures 2.1b and 2.1c illustrate the atomic orbital energy dependence on the principal quantum number and proton-electron radius, respectively.

When a single H atom is optically excited, the electron is promoted from the lowest lying

1s orbital into a higher lying orbital. The electronic transition is described by:

$$\langle \Psi_f(r) | \hat{\mu}_{if}(r) | \Psi_i(r) \rangle = \int \Psi_f^*(r) \,\hat{\mu}_{if}(r) \,\Psi_i(r) \,d^3r$$
 (2.4)

where $\Psi_i(r)$ and $\Psi_f(r)$ are the wavefunctions of the initial and final states and $\hat{\mu}_{if}(r)$ is the transition dipole moment between them. In order for the integral to be non-zero, the initial and final state wavefunctions must be a pair of even and odd functions. This optical selection rule allows only transitions between atomic orbitals of even and odd azimuthal (i.e. orbital angular momentum) quantum number ℓ to occur.

2.2 The Hydrogen Molecule

In the simplest molecular case, where two protons are bound by one electron, the electronic Hamiltonian of the H_2^+ molecule in atomic units and under the Born-Oppenheimer approximation takes the form:

$$\hat{H}_{el}\left(\overrightarrow{R}_{a},\overrightarrow{R}_{b},\overrightarrow{r}\right) = -\frac{1}{2}\nabla_{r}^{2} - \frac{1}{\left|\overrightarrow{R}_{a}-\overrightarrow{r}\right|} - \frac{1}{\left|\overrightarrow{R}_{b}-\overrightarrow{r}\right|} + \frac{1}{\left|\overrightarrow{R}_{a}-\overrightarrow{R}_{b}\right|}$$
(2.5)

where \overrightarrow{r} is the dynamic position of the electron and \overrightarrow{R}_i is the fixed position of the hydrogen atom H_i , illustrated in figure 2.2a [87]. The first term in equation 2.5 is the electron kinetic energy, the second and third terms are the attractive electron-proton Coulombic interactions, and the final term is the repulsive proton-proton Coulombic interaction. The second, third, and fourth terms are collectively called the potential energy \hat{V}_{ab} . The Schrödinger equation for the electronic Hamiltonian of the H_2^+ atom, $\hat{H}_{el}\left(\overrightarrow{R}\right)\Psi\left(\overrightarrow{r},\overrightarrow{R}\right) = E_{el}\left(\overrightarrow{R}\right)\Psi\left(\overrightarrow{r},\overrightarrow{R}\right)$, is dependent on the dynamic position of the electron and fixed positions of the nuclei. Solving the Schrödinger equation, the lowest lying molecular orbitals are linear combinations of the atomic orbitals of hydrogen, illustrated in figure 2.2c:



Figure 2.2: Hydrogen H_2^+ molecular orbitals. (a) Hydrogen molecule diagram, where p_i^+ and e^- are the protons and electron, respectively, and \overrightarrow{R}_i and \overrightarrow{r} are their positions, respectively. (b) Molecular orbital diagram where the 1s σ bonding and 1s σ^* anti-bonding molecular orbitals are linear combinations of the two 1s hydrogen atomic orbitals. (c) H 1s atomic orbitals and H_2^+ 1s molecular bonding and anti-bonding orbitals.

$$\Psi_{\pm}\left(\overrightarrow{r},\overrightarrow{R}\right) = c_a\left(\overrightarrow{R}\right)\phi_a\left(\overrightarrow{r},\overrightarrow{R}\right) \pm c_b\left(\overrightarrow{R}\right)\phi_b\left(\overrightarrow{r},\overrightarrow{R}\right)$$
(2.6)

where $\Psi_{\pm}\left(\overrightarrow{r}, \overrightarrow{R}\right)$ are the bonding and anti-bonding molecular orbitals, $c_a = c_b = 1/\sqrt{2}$ are the coefficients describing the linear combination, and $\phi_i\left(\overrightarrow{r}, \overrightarrow{R}\right)$ are the atomic orbitals of H_i . The energies of the molecular orbitals, illustrated in figure 2.2b, depend on the positions of the hydrogen atoms:

$$E_{\pm}\left(\overrightarrow{R}\right) = -\frac{E\left(\overrightarrow{R}\right) \pm V_{ab}\left(\overrightarrow{R}\right)}{1 \pm S_{ab}\left(\overrightarrow{R}\right)}$$
(2.7)

$$E\left(\overrightarrow{R}\right) = \left\langle \phi\left(\overrightarrow{r}, \overrightarrow{R}\right) \left| \hat{H}_{el}\left(\overrightarrow{r}, \overrightarrow{R}\right) \right| \phi\left(\overrightarrow{r}, \overrightarrow{R}\right) \right\rangle$$
$$= \int \phi^*\left(\overrightarrow{r}, \overrightarrow{R}\right) \hat{H}_{el}\left(\overrightarrow{r}, \overrightarrow{R}\right) \phi\left(\overrightarrow{r}, \overrightarrow{R}\right) d^3\overrightarrow{r}$$
(2.8)

$$V_{ab}\left(\overrightarrow{R}\right) = \left\langle \phi_b\left(\overrightarrow{r},\overrightarrow{R}\right) \left| \hat{V}_{ab}\left(\overrightarrow{r},\overrightarrow{R}\right) \right| \phi_a\left(\overrightarrow{r},\overrightarrow{R}\right) \right\rangle$$
$$= \int \phi_b^*\left(\overrightarrow{r},\overrightarrow{R}\right) \hat{V}_{ab}\left(\overrightarrow{r},\overrightarrow{R}\right) \phi_a\left(\overrightarrow{r},\overrightarrow{R}\right) d^3\overrightarrow{r}$$
(2.9)

$$S_{ab}\left(\overrightarrow{R}\right) = \left\langle \phi_b\left(\overrightarrow{r}, \overrightarrow{R}\right) \middle| \phi_a\left(\overrightarrow{r}, \overrightarrow{R}\right) \right\rangle$$
$$= \int \phi_b^*\left(\overrightarrow{r}, \overrightarrow{R}\right) \phi_a\left(\overrightarrow{r}, \overrightarrow{R}\right) d^3\overrightarrow{r}$$
(2.10)

where $E\left(\overrightarrow{R}\right)$ is the atomic orbital energy, $V_{ab}\left(\overrightarrow{R}\right)$ is the expectation value of the electron potential energy, and $S_{ab}\left(\overrightarrow{R}\right)$ is the atomic orbital overlap of the hydrogen atoms $H_a(\overrightarrow{R}_a)$ and $H_b(\overrightarrow{R}_b)$. Optical excitations in molecules are described identically to those in atoms:

$$\left\langle \Psi_f\left(\overrightarrow{r},\overrightarrow{R}\right) \left| \hat{\mu}_{if}\left(\overrightarrow{r},\overrightarrow{R}\right) \right| \Psi_i\left(\overrightarrow{r},\overrightarrow{R}\right) \right\rangle$$
 (2.11)

where the transition dipole moment $\hat{\mu}_{if}\left(\overrightarrow{r}, \overrightarrow{R}\right)$ connects the initial and final molecular orbitals $\Psi_i\left(\overrightarrow{r}, \overrightarrow{R}\right)$ and $\Psi_f\left(\overrightarrow{r}, \overrightarrow{R}\right)$. Thus, the same optical selection rule applies and only allows transitions between molecular orbitals of different symmetry to occur.

2.3 Solid State Materials

In a crystalline solid, the electronic structure is described by many electrons in a periodic potential, illustrated in figure 2.3a [88]. The Hamiltonian of a crystalline solid takes the form:

$$\hat{H} = -\sum_{i} \frac{\hbar^2}{2m_e} \nabla_i^2 + \sum_{i} U\left(\overrightarrow{r}_i\right) + \sum_{i < j} \frac{e^2}{\left|\overrightarrow{r}_i - \overrightarrow{r}_j\right|}$$
(2.12)

where the first term describes the electron kinetic energies and \hbar is Planck's constant and m_e the electron mass, the second term the attractive electron-nuclei Coulombic interactions and \overrightarrow{r}_i are the electron positions, and the third term the repulsive electron-electron Coulombic interactions and e is the electron charge. The wavefunctions take the form of



Figure 2.3: Solid state wavefunctions. (a) Solid state diagram in one dimension, where N^+ and e^- are the nuclei and electron, respectively, and \overrightarrow{R} and \overrightarrow{r} are their positions, respectively. (b) Nuclei atomic orbitals and solid state "bonding" and "anti-bonding" orbitals.

anti-symmetrized products of the localized atomic orbitals ϕ , shown in figure 2.3b:

$$\Psi = \hat{\alpha} \prod_{i} \phi \left(\overrightarrow{r}_{i} - \overrightarrow{R}_{i} \right)$$
(2.13)

where $\hat{\alpha}$ is the anti-symmetrization operator and \overrightarrow{R}_i are the nuclei positions.

The full Hamiltonian of a crystalline solid is extremely complicated. However, the eigenstates and eigenvalues can be calculated using a periodic, one-electron Hamiltonian

$$\hat{H} = \frac{\hbar^2}{2m_e} \nabla^2 + U\left(\overrightarrow{r}\right) \tag{2.14}$$

where $U(\vec{r}) = U(\vec{r} + \vec{R})$ is the periodic potential defined by the lattice vector \vec{R} (i.e. the inter-unit cell distance). In this method, the repulsive electron-electron interactions described in the second term of equation 2.12 must be added back in to accurately described the electron dynamics. The one-electron eigenstates are products of a plane wave and periodic function

as described by Bloch's theorem

$$\psi_{nk}\left(\overrightarrow{r}\right) = \exp\left[i\overrightarrow{k}\cdot\overrightarrow{r}\right]u_{nk}\left(\overrightarrow{r}\right)$$
(2.15)

where $\overrightarrow{k} = 2\pi/\overrightarrow{R}$ is the wavevector describing the periodicity of the crystal lattice and u_{nk} is a periodic function that takes the form of the contributing atomic orbitals of each nucleus. Due to the periodicity of the crystal lattice, the eigenstates have the special property that $\psi\left(\overrightarrow{r}+\overrightarrow{R}\right) = \exp\left[i\overrightarrow{k}\cdot\overrightarrow{R}\right]\psi(\overrightarrow{r}).$

The Schrödinger equation of a crystalline solid thus reduces to $\hat{H}u_{nk}(\vec{r}) = E_{nk}u_{nk}(\vec{r})$. Like the eigenstates, the periodic function also has the special property that $u_{nk}(\vec{r}) = u_{nk}(\vec{r} + \vec{R})$. In the solid state, n is a discrete quantum number describing the electronic state and \vec{k} is a continuous quantum number describing the crystal momentum. The eigenvalues of a solid state material become continuous electronic energy bands dependent on the crystal lattice structure and elemental composition of the material.

As in the case of atoms and molecules, the same optical selection rule continues to apply and only allows transitions between electronic bands of different symmetry.

2.3.1 Semiconductors

In a semiconductor, low lying electronic bands are completely filled and there is an energy gap between the highest lying, occupied valence band and the lowest lying, unoccupied conduction band. The band gap energy is typically less than or equal to 2 eV. As the band gap energy increases, the material becomes an insulator and as the band gap energy goes to zero (i.e. incomplete filling of an electronic band), the material becomes a metal. The band gap can either be direct or indirect, where the highest lying valence band maximum and lowest lying conduction band minimum have the same or different wavevector k, respectively. In general, direct band gap semiconductors are substantially more optically active than indirect semiconductors because the latter require phonons play a role in excitation and emission processes in order to conserve crystal momentum.

Optical excitation of a direct band gap semiconductor promotes an electron from a filled valence band into an unoccupied conduction band. The promotion of a negatively charged electron into the conduction band results in a positively charged hole in the valence band due to the absence of the excited electron. Repulsion of the conduction band electron by valence band electrons localizes the excited electron near the valence band hole. Coulombic attraction between the electron and hole can bind them to form a charge neutral (bosonic) quasiparticle called an exciton, illustrated in figures 2.4a and b [89]. The energy difference between the free electron and hole states and the exciton state is called the exciton binding energy: $\Delta E_{Ex} = (E_e + E_h) - E_{Ex}$, and is typically on the order of $0.01 - 1 \, eV$.

Conduction band electrons and valence band holes tend to be localized near the minimum and maximum of their respective bands, at which point the relationship between energy and wavevector (i.e. the dispersion relation) is approximately quadratic. Thus, the electronic band structure of the electron and hole can be written as:

$$E_{e}(k) = E_{c} + \frac{\hbar^{2}k^{2}}{2m_{e}}$$
(2.16)

$$E_h(k) = E_v + \frac{\hbar^2 k^2}{2m_h}$$
 (2.17)

where E_c and E_v are the minimum and maximum energies of the conduction and valence bands, respectively, and m_e and m_h are the electron and hole masses, respectively. A qualitative picture of the electronic band structure and exciton state is illustrated in figure 2.4b.

Intuitively, excitons can be thought of as hydrogen-like atoms, where the negatively charged electron orbits around the positively charged hole. However, there are two major differences between an exciton and a hydrogen atom. First, the electron and hole masses are comparable in magnitude, unlike atoms where there is a three order of magnitude disparity between the proton and electron masses. Second, excitons are described by the dielectric constant of the solid state material (i.e. relative permittivity), which screens the electron-



Figure 2.4: Exciton band structure. (a) Exciton diagram, where h^+ and e^- are the hole and electron and \overrightarrow{r} is the distance between them. (b) Qualitative exciton band structure with a free electron in the conduction band (blue) and a hole in the valence band (green), where E_{BG} is the direct band gap energy. Electrostatic repulsion of the free electron in the conduction band by the bound electrons in the valence band localize the electron and hole near each other. The electron and hole bind due to electrostatic attraction into an exciton (purple) and are stabilized by the exciton binding energy ΔE_{Ex} . (c) Frenkel excitons are characterized by small electron-hole (green and blue, respectively) radii and large exciton binding energies. (d) Wannier-Mott excitons are characterized by large electron-hole radii and small exciton binding energies.

hole interaction, rather than in a non-perturbative vacuum. Nevertheless, the similarities between excitons and hydrogen atoms are quite striking. Similar to hydrogen atoms, excitons possess Rydberg states where the energy is dependent on the quantum number n and are given by:

$$E_{Ex}^{n}(k) = E_{BG}(0) + \frac{\hbar^{2}k^{2}}{2(m_{e} + m_{h})} - \frac{R_{y}^{*}}{n^{2}}$$
$$= (E_{e}(0) - E_{h}(0)) + \frac{\hbar^{2}k^{2}}{2(m_{e} + m_{h})} - \left(\frac{\varepsilon_{0}}{\varepsilon}\right)^{2} \left(\frac{m^{*}}{m_{e}}\right) \frac{R_{y}}{n^{2}}$$
(2.18)

where the first term $E_{BG}(0)$ is the band gap energy, the second term describes the electron and hole kinetic energy given by the dispersion relation, and the third term is the exciton Rydberg energy given by the exciton Rydberg constant R_y^* and composed of the relative permittivity of the material $(\varepsilon/\varepsilon_0)$ and the electron-hole reduced mass $m^* = m_e m_h/(m_e + m_h)$. In this case, the first exciton Rydberg term (n = 1) gives the exciton binding energy.

$$\Delta E_{Ex} = R_y^* = \left(\frac{\varepsilon_0}{\varepsilon}\right)^2 \left(\frac{m^*}{m_e}\right) R_y \tag{2.19}$$

An exciton Bohr radius can also be defined and describes the exciton delocalization or most probable distance between the electron and hole in the lowest lying 1s exciton state.

$$a_{Ex} = \frac{\epsilon}{\epsilon_0} \frac{m_e}{m^*} a_0 \tag{2.20}$$

2.3.1.1 Frenkel Excitons

In semiconductors with small dielectric constants, electrons and holes experience little screening from each other and tend to interact strongly. In this case, the quasiparticles are classified as Frenkel excitons, shown in figure 2.4c. The delocalization of Frenkel excitons tends to be small (on the order of a unit cell) and the exciton binding energy large (0.1 - 1 eV). Examples of Frenkel excitons include those in alkali halides [90], organic molecular crystals [91], and H- and J-aggregates [38].

2.3.1.2 Wannier-Mott Excitons

In semiconductors with large dielectric constants, electrons and holes are substantially screened from each other and tend to weakly interact. In this case, the quasiparticles are classified as Wannier-Mott excitons, shown in figure 2.4d. The delocalization of Wannier-Mott excitons tends to be large (greater than a unit cell) and the exciton binding energy small (0.01 - 0.1 eV). Examples of Wannier-Mott excitons include many inorganic semiconductors including bulk group IV silicon [92] and germanium [93], III-V compounds such as pseudo two-dimensional GaAs quantum wells [94], and II-VI compounds such as ZnO nanowires [95] and CdSe quantum dots [96].

2.3.1.3 Intermediate Excitons

It is important to note that Frenkel and Wannier-Mott excitons are only the extrema of a spectrum of excitons. Many systems, such as single-walled carbon nanotubes [97], lie within this spectrum. In the case of single-walled carbon nanotubes, excitons are delocalized over several nanometers due to the large dielectric constant but screening is reduced in low dimensions and allows for large exciton binding energies. Thus, excitons in single-walled carbon nanotubes possess both Frenkel and Wannier-Mott characteristics and cannot be classified as either.

2.4 Many-Body Interactions

Expanding the analogy of excitons to hydrogen-like atoms, excitons also form the basis of more complex excitonic "ions" and "molecules". Here, we will discuss higher-order correlations of excitons and charged particles that result from n > 2 particle interactions.

2.4.1 Trions: Excitonic Ions

Similar to atoms, excitons can form positively and negatively charged "ions" [98, 99]. Threeparticle correlations of an exciton with an additional electron or hole produce positively and negatively charged excitons called trions, which are analogous to H_2^+ and H^- , respectively. A trion binding energy and Bohr radius can be defined to describe the degree of stabilization and delocalization. However, since trions are not produced directly by optical excitation, we do not observe these quasi-particles in our spectroscopic experiments.

2.4.2 Multi-Excitons: Excitonic Molecules

Similar to atoms, excitons can form excitonic "molecules" [61, 62, 100]. The simplest case is the four-particle correlation of identical electrons and holes called a biexciton, which is analogous to a homonuclear diatomic molecule (e.g. H_2). A four-particle correlation of nonequivalent electrons and holes can form called a mixed biexciton, which is analogous to a heteronuclear diatomic molecule (e.g. HCl). A biexciton binding energy can be defined to describe the degree of exciton-exciton stabilization:

$$\Delta E_{Bx}^{nm} = (E_{Ex}^n + E_{Ex}^m) - E_{Bx}^{nm} \tag{2.21}$$

where E_{Bx}^{nm} is the energy of the biexciton composed of n and m excitons and E_{Ex}^{n} and E_{Ex}^{m} are the constituent n and m excitons energies. Like electrons in a molecular bond, the constituent exciton spins of a biexciton must be anti-parallel $(|\uparrow,\downarrow\rangle \text{ or } |\downarrow,\uparrow\rangle)$ in order to bind. Excitons with parallel spins $(|\uparrow,\uparrow\rangle \text{ or } |\downarrow,\downarrow\rangle)$ can be correlated to each other but do not exhibit a stabilizing binding energy due to the Pauli exclusion principle. Higher-order correlations of excitons can also form called multi-excitons, which are analogous to polyatomic molecules.

In our spectroscopic experiment, multi-excitons can be generated via successive electromagnetic field interactions up and down the ladder of n-exciton correlations. In our twodimensional spectrometer, the phase stability amongst an arbitrary number and geometry
of pulses provides a unique capability that allows us to investigate this ladder of countable n-exciton correlations unlike most conventional two-dimensional spectrometers.

2.5 Exciton-Phonon Coupling

In the solid state, electronic structure is strongly dependent on the structure of the material. Thus, deformations in the crystal structure via collective motions of atoms or molecules in the lattice (i.e. phonons) can strongly couple to the electronic modes and modulate their eigenstates and eigenvalues [101]. This exciton-phonon coupling can be described by adding an exciton-phonon interaction term \hat{V}_{EP} to the Hamiltonian:

$$\hat{H} = \hat{H}_E + \hat{H}_P + \hat{V}_{EP}$$

$$= \sum_{n,k} a^{\dagger}_{n,k} a_{n,k} E_n\left(\overrightarrow{k}\right) + \sum_{m,q} b^{\dagger}_{m,q} b_{m,q} E_m\left(\overrightarrow{q}\right) + \hat{V}_{EP}$$
(2.22)

$$\hat{V}_{EP} = \sum_{f,i} \sum_{k',k} \sum_{m,q} V_{fk',ik,mq} \left| \Psi_f\left(\overrightarrow{k'}\right) \right\rangle \left\langle \Psi_i\left(\overrightarrow{k}\right) \right| \left(b_{m,q}^{\dagger} + b_{m,q} \right)$$
(2.23)

where $a_{n,k}$ and $a_{n,k}^{\dagger}$ are the creation and annihilation operators of exciton n with wavevector \overrightarrow{k} and $b_{m,q}$ and $b_{m,q}^{\dagger}$ are the creation and annihilation operators of phonon m with wavevector \overrightarrow{q} . $V_{fk',ik,mq}$ is the coupling strength between phonon mode m and exciton i that results in exciton f, where $\overrightarrow{k'} = \overrightarrow{k} + \overrightarrow{q}$. This linear coupling term allows the creation or annihilation of a single phonon upon an electronic transition. Strong coupling between exciton and phonon modes will be discussed further in chapter 7 in our analysis of the biexciton states in cuprous oxide and chapter 8 in our discussion of monolayer transition metal dichalcogenides.

2.6 Excitons on the Nanoscale

When the spatial dimensions of a nanostructured semiconductor are on the order of the exciton de Broglie wavelength $\lambda = h/p$ (approximately 10 nm at room temperature), the



Figure 2.5: Particle-in-a-box. (a) Particle-in-a-box potential well and wavefunctions n = 1 - 5. (b) Eigenstate energies as a function of quantum number n with, where $L = a_0$. (c) Spatial dependence of eigenstate energies on the width of the potential L, where n = 1.

exciton wavefunctions are perturbed due to constraint by the size of the material. First, we will introduce the affects of spatial confinement by considering the easiest case of a particlein-a-box. Second, we will discuss additional affects of spatial confinement in nanostructured semiconductors in various dimensions.

2.6.1 Particle-in-a-Box

In the case of atoms and excitons, the Bohr radius defines the most probable distance between the positively and negatively charged particles. When a particle is spatially confined on the order of its Bohr radius, the eigenstates and eigenvalues of the system are perturbed [87]. Consider a particle in a one dimensional box of length L with zero potential energy inside and infinite potential energy outside the box, illustrated in figure 2.5a.

$$\hat{V}(x) = \begin{cases}
\infty, & x < -L/2 \\
0 & -L/2 < x < L/2 \\
\infty, & x > L/2
\end{cases}$$
(2.24)

The Hamiltonian thus takes the form:

$$\hat{H} = -\frac{\hbar^2}{2m}\nabla^2 + \hat{V}(x)$$
(2.25)

where *m* is the mass of the particle. Solving the Schrödinger equation $\hat{H}\Psi(x) = E\Psi(x)$ produces wavefunctions that are only defined inside the box, illustrated in figure 2.5a.

$$\Psi(x) = \begin{cases} 0, & x < -L/2 \\ \sin\left[k_n\left(x + \frac{L}{2}\right)\right], & -L/2 < x < L/2 \\ 0, & x > L/2 \end{cases}$$
(2.26)

where the spatial wavevector $k_n = n\pi/L$ defines the periodicity of the wavefunction in the box. The particle-in-a-box energies thus take the form:

$$E_n = \frac{\hbar^2 k_n^2}{2m} = \frac{\pi^2 \hbar^2 n^2}{2mL^2}$$
(2.27)

The energy levels of the particle-in-a-box states are similar to the exciton Rydberg state energies, however there is a striking difference. The particle-in-a-box energies increase quadratically with the quantum number n (figure 2.5b), while the exciton energies are inversely proportional to the second power of n (figure 2.1b). While the exciton energy levels get closer together with increasing n, the particle-in-a-box energy levels get farther apart due to greater confinement of larger electron orbitals. However, the particle-in-a-box and exciton energies are both inversely proportional to the second power on the spatial localization, Land a_0 , respectively. Thus, both the particle-in-a-box and exciton states are stabilized by



Figure 2.6: Quantum confinement. The change in the density of states with decreasing dimensionality. Oscillator strength is concentrated into narrow regions of the electromagnetic spectrum due to the increase in the crystal momentum uncertainty and a flattening of the electronic bands.

spatial delocalization.

2.6.2 Quantum Confinement

From the discussion of a particle-in-a-box, we learn that exciton energies increase as spatial confinement increases. Additionally, the oscillator strengths of the associated electronic transitions are concentrated into narrow regions of the electromagnetic spectrum, enhancing absorption and emission [102]. The oscillator strength concentration can be understood by considering Heisenberg's uncertainty principle: $\sigma_x \sigma_p \geq \hbar/2$. For an exciton in the bulk, the uncertainty in the crystal momentum (σ_p) is small, unlike the uncertainty in the exciton position (σ_x), due to the well defined periodicity and symmetry of the crystal. However for a spatially confined exciton, the uncertainty in the exciton position decreases as confinement increases and requires an increase in the crystal momentum uncertainty. The decrease in the crystal momentum uncertainty results in a flattening of the electronic bands and a concentration in the density of states, illustrated in figure 2.6. The role of quantum confinement will be discussed further in chapter 8 in our measurements of the single and multi-exciton states of monolayer transition metal dichalcogenides.

Chapter 3

Third-Order Nonlinear Spectroscopy

In chapter 3, we will describe the spectroscopic techniques used to study excitons in nanostructured materials. First, we will begin our discussion of spectroscopy with linear absorption and the first-order response function. From there, we will generalize our discussion to third-order response functions in order to understand the theoretical principles of third-order non-linear spectroscopy. We will discuss pump-probe, transient grating, and two-dimensional spectroscopies, which have increasing numbers of experimental variables and allow new spectroscopic signals to be measured. We will also comment on higher-order two-dimensional spectroscopy. Finally, we will end by discussing excitation-induced spectroscopic artifacts.

For the theoretical description of nonlinear spectroscopy, there are a great number of texts that help form the foundation of this discussion and to which I refer the reader for additional information [41, 103, 104, 105].

Spectroscopy is the study of physical phenomena by the interaction of light and matter [106]. First, we recognize that matter is composed of charged particles (i.e. protons and electrons) and light is an electromagnetic field which can exert a force on charged particles. In the classical picture, an electromagnetic field exerts a force that depends on the form of the binding potential for the charged particles, the charge magnitudes, and the form and

amplitude of the electromagnetic field. It is this exerted force that allows light and matter to interact. Moreover, the acceleration of charged particles due to exerted electromagnetic forces allows matter to radiate new electromagnetic fields that form the heart of spectroscopy.

3.1 Linear Absorption Spectroscopy

In linear absorption, an electromagnetic field creates a macroscopic oscillating polarization as it propagates through a material:

$$P^{(1)}\left(\omega, \overrightarrow{k}\right) = \chi^{(1)}\left(\omega, \overrightarrow{k}\right) E_{I}^{(1)}\left(\omega, \overrightarrow{k}\right)$$
(3.1)

where $P^{(1)}\left(\omega, \vec{k}\right)$ is the first-order (linear) polarization as a function of frequency ω and wavevector \vec{k} , $\chi^{(1)}\left(\omega, \vec{k}\right)$ is the first-order susceptibility of the material, and $E_I^{(1)}\left(\omega, \vec{k}\right)$ is the spectral density of the incident electromagnetic field. The gradient of the generated linear polarization radiates a second electromagnetic field:

$$E_R^{(1)}\left(\omega, \overrightarrow{k}\right) = iP^{(1)}\left(\omega, \overrightarrow{k}\right) \tag{3.2}$$

that is dependent on the electronic resonances of the material. This radiating electromagnetic field is $\frac{\pi}{2}$ out of phase with the incident field and destructively interferes at the resonance energies of the material. The total electromagnetic field can be written as

$$E_{tot}^{(1)}\left(\omega, \overrightarrow{k}\right) = E_{I}^{(1)}\left(\omega, \overrightarrow{k}\right) + E_{R}^{(1)}\left(\omega, \overrightarrow{k}\right)$$
$$= E_{I}^{(1)}\left(\omega, \overrightarrow{k}\right) \left[1 + i\chi^{(1)}\left(\omega, \overrightarrow{k}\right)\right]$$
(3.3)

assuming only a small change in the amplitude of the incident electromagnetic field. However, the intensity rather than the amplitude of the electromagnetic field is measured by a



Figure 3.1: Linear absorption experiment. (a) Laser beam geometry. (b) Experimental diagram. (c) Energy level diagram. (d) Feynman diagram.

spectrometer. So, the measured spectral output takes the form

$$I_{tot}^{(1)}\left(\omega, \overrightarrow{k}\right) = \left|E_{tot}^{(1)}\left(\omega, \overrightarrow{k}\right)\right|^{2}$$
$$= I_{I}^{(1)}\left(\omega, \overrightarrow{k}\right)\left|1 + \chi^{(1)}\left(\omega, \overrightarrow{k}\right)\right|^{2}$$
(3.4)

Noting the susceptibility is a complex number $\chi^{(1)} = \chi_{\Re}^{(1)} + i\chi_{\Im}^{(1)}$ and assuming the susceptibility is small compared to the incident electromagnetic field $\chi^{(1)} \ll 1$, equation 3.4 reduces to

$$I_{tot}^{(1)}\left(\omega,\overrightarrow{k}\right) = I_{I}^{(1)}\left(\omega,\overrightarrow{k}\right) \left[1 - 2\chi_{\Im}^{(1)}\left(\omega,\overrightarrow{k}\right)\right]$$
(3.5)

In this case, the $1 - 2\chi_{\Im}^{(1)}(\omega, \vec{k})$ term in 3.5 equation describes the destructive interference of the two electromagnetic fields and the imaginary component of the susceptibility describes the absorptive response of the material to the incident electromagnetic field. A linear absorption experiment is graphically illustrated in figure 3.1.

The susceptibility can also be recast in the time domain. The Fourier transform of the first-order susceptibility is called the linear response function:

$$R^{(1)}\left(t,\overrightarrow{k}\right) = \int_{-\infty}^{\infty} \chi^{(1)}\left(\omega,\overrightarrow{k}\right) \exp\left[-i2\pi\omega t\right] d\omega$$
(3.6)

The linear polarization generated in the material can also be recast in the time domain with

the temporal profile of the incident electromagnetic field $E_I^{(1)}\left(t, \overrightarrow{k}\right)$:

$$P^{(1)}\left(t,\overrightarrow{k}\right) = \int_{-\infty}^{t} R^{(1)}\left(t-\tau,\overrightarrow{k}\right) E_{I}^{(1)}\left(\tau,\overrightarrow{k}\right) d\tau \qquad (3.7)$$

In the time domain, the linear polarization generated is a convolution of the response function with the incident electromagnetic field; while in the frequency domain, the polarization generated is the product of the susceptibility and the incident electromagnetic field. The polarization generated by the incident electromagnetic field acts as a source term that can be inserted into Maxwell's equations:

$$\nabla^{2} E\left(t, \overrightarrow{k}\right) - \frac{1}{c^{2}} \frac{\partial^{2} E\left(t, \overrightarrow{k}\right)}{\partial t^{2}} = \frac{4\pi}{c^{2}} \frac{\partial^{2} P^{(1)}\left(t, \overrightarrow{k}\right)}{\partial^{2} t}$$
(3.8)

and produces solutions for the polarization of the form

$$P^{(1)}\left(t,\overrightarrow{k}\right) = P^{(1)}\left(t\right)\exp\left[i\overrightarrow{k}\cdot\overrightarrow{r} - i\omega t\right] + c.c.$$
(3.9)

where $P^{(1)}(t)$ is an envelope function and radiates an electric field of the same form

$$E^{(1)}\left(t,\overrightarrow{k}\right) = E^{(1)}\left(t\right)\exp\left[i\overrightarrow{k}\cdot\overrightarrow{r} - i\omega t\right] + c.c.$$
(3.10)

Figure 3.2 illustrates an incident laser field, material resonance, and signal transmission in the time and frequency domains.

In the time domain, the linear polarization can be calculated using the transition dipole moment operator $\overrightarrow{\mu}(t)$ and the time-dependent density matrix $\rho(t)$:

$$P^{(1)}(t) = Tr\left(\overrightarrow{\mu}(t)\rho(t)\right) \tag{3.11}$$

where Tr is the trace of a matrix. The density matrix can be calculated from the von



Figure 3.2: Linear absorption spectrum. (a) Incident laser spectrum. (b) Material absorption resonance. (c) Laser spectrum transmission. (d) Incident laser field. (e) Emitted time domain signal by the material. (f) Total time domain signal.

Neumann equation:

$$\rho(t) = -\frac{i}{\hbar} \int_{-\infty}^{t} dt_1 \left[\hat{V}(t_1), \rho(0) \right]$$
(3.12)

where [a, b] = ab - ba is the commutator operator and $\hat{V}(t) = -\overrightarrow{\mu}(t) \cdot \overrightarrow{E}(t)$ is the timedependent component of the Hamiltonian, $\hat{H}(t) = \hat{H}_0 + \hat{V}(t)$, describing the light-matter interaction. Substituting equation 3.12 into equation 3.11 and simplifying, we find the linear polarization can be written as

$$P^{(1)}(t) = \frac{i}{\hbar} \int_0^\infty d\tau \, E\left(t - \tau\right) Tr\left(\left[\overrightarrow{\mu}(\tau), \overrightarrow{\mu}(0)\right]\rho(0)\right) \tag{3.13}$$

with the change of variable $\tau = t - t_1$. Since the linear polarization is a convolution of the response function with the incident electromagnetic field, we observe that the linear response function is

$$R^{(1)}(\tau) = \frac{i}{\hbar} \theta(\tau) Tr\left(\left[\overrightarrow{\mu}(\tau), \overrightarrow{\mu}(0)\right] \rho(0)\right)$$
(3.14)

where $\theta(t)$ is the Heaviside step function.

Both the frequency and time domains are useful paradigms from which to understand spectroscopy. We will use both perspectives to think about nonlinear spectroscopy and understand the time-resolved experiments described in this thesis.

3.2 Third-Order Nonlinear Spectroscopy

With an understanding of first-order (linear) spectroscopy, we will now discuss third-order nonlinear spectroscopy. In third-order spectroscopy, we consider the case where there are three light matter interactions. In this case, the principles of linear spectroscopy are easily generalizable to higher-order. First, the polarization generated in a material by an electromagnetic field(s) can be expanded to higher-order:

$$P(\omega) = P^{(0)}(\omega) + P^{(1)}(\omega) + P^{(2)}(\omega) + P^{(3)}(\omega) + \cdots$$
(3.15)

$$P(t) = P^{(0)}(t) + P^{(1)}(t) + P^{(2)}(t) + P^{(3)}(t) + \cdots$$
(3.16)

In the frequency domain, the third-order polarization $P^{(3)}(\omega)$ in equation 3.15 can easily be written down as the product of the third-order susceptibility $\chi^{(3)}(\omega)$ and the spectral densities of the three incident electromagnetic fields:

$$P^{(3)}(\omega) = \chi^{(3)}(\omega) E_1(\omega) E_2(\omega) E_3(\omega)$$
(3.17)

In the time domain, the third-order polarization $P^{(3)}(t)$ in equation 3.16 is more difficult to write down as it requires the calculation of the third-order expansion of the density matrix $\rho^{(3)}(t)$ and results in a triple integral and a triple nested commutator:

$$P^{(3)}(t) = Tr\left(\overrightarrow{\mu}(t)\rho^{(3)}(t)\right)$$
(3.18)

$$\rho^{(3)}(t) = \left(-\frac{i}{\hbar}\right)^3 \int_{-\infty}^t dt_3 \int_{-\infty}^{t_3} dt_2 \int_{-\infty}^{t_2} dt_1 \left[-\overrightarrow{\mu}(t_3) \cdot \overrightarrow{E_3}(t_3), \left[-\overrightarrow{\mu}(t_2) \cdot \overrightarrow{E_2}(t_2), \left[\overrightarrow{\mu}(t_1) \cdot \overrightarrow{E_1}(t_1), \rho(0)\right]\right]\right] \quad (3.19)$$

With the third-order expansion of the density matrix, we can write down the full third-order polarization:

$$P^{(3)}(t') = \left(\frac{i}{\hbar}\right)^3 \int_0^\infty dt \int_0^\infty dT \int_0^\infty d\tau \,\overrightarrow{E_3}(t'-t) \,\overrightarrow{E_2}(t'-t-T) \,\overrightarrow{E_1}(t'-t-T-\tau) \\ \times Tr\left(\left[\left[\overrightarrow{\mu}(\tau+T+t), \overrightarrow{\mu}(\tau+T)\right], \overrightarrow{\mu}(\tau)\right], \overrightarrow{\mu}(0)\right] \rho(0)\right) \quad (3.20)$$

with the change of variables: $t_1 = t' - \tau - T - t$, $t_2 = t' - T - t$, $t_3 = t' - t$ and enforcing time ordering of the pulses $t_1 \leq t_2 \leq t_3$. Since the third-order polarization is a triple convolution of the response function with three incident electromagnetic fields, we observe from equation 3.20 that the third-order response function is

$$R^{(3)}(\tau, T, t) = \left(\frac{i}{\hbar}\right)^{3} \theta(t) \theta(T) \theta(\tau) Tr\left(\left[\left[\overrightarrow{\mu}(\tau + T + t), \overrightarrow{\mu}(\tau + T)\right], \overrightarrow{\mu}(\tau)\right], \overrightarrow{\mu}(0)\right] \rho(0)\right)$$
(3.21)

The triple nested commutator in equation 3.21 results in the sum of eight correlation functions at third-order: four are unique and the other four are their complex conjugates. The four unique correlation functions can be written as:

$$R^{(3)}(\tau, T, t) = \left(\frac{i}{\hbar}\right)^{3} \theta(t) \theta(T) \theta(\tau) \sum_{i=1}^{4} \left[R_{i}^{(3)}(t, T, \tau) - R_{i}^{(3)*}(t, T, \tau)\right]$$
(3.22)



Figure 3.3: One-quantum Feynman diagrams. (a) Unique correlation functions of the thirdorder response function $R^{(3)}(\tau, T, t)$. (b) Two-level system energy level energy diagram.

$$R_{1}^{(3)}\left(\tau+T+t\right) = Tr\left[\overrightarrow{\mu}\left(\tau+T+t\right)\overrightarrow{\mu}\left(\tau+T\right)\overrightarrow{\mu}\left(\tau\right)\overrightarrow{\mu}\left(0\right)\rho\left(0\right)\right]$$
(3.23)

$$R_2^{(3)}\left(\tau + T + t\right) = Tr\left[\overrightarrow{\mu}\left(0\right)\overrightarrow{\mu}\left(\tau + T\right)\overrightarrow{\mu}\left(\tau + T + t\right)\overrightarrow{\mu}\left(\tau\right)\rho\left(0\right)\right]$$
(3.24)

$$R_{3}^{(3)}\left(\tau+T+t\right) = Tr\left[\overrightarrow{\mu}\left(0\right)\overrightarrow{\mu}\left(\tau\right)\overrightarrow{\mu}\left(\tau+T+t\right)\overrightarrow{\mu}\left(\tau+T\right)\rho\left(0\right)\right]$$
(3.25)

$$R_4^{(3)}\left(\tau + T + t\right) = Tr\left[\overrightarrow{\mu}\left(\tau\right)\overrightarrow{\mu}\left(\tau + T\right)\overrightarrow{\mu}\left(\tau + T + t\right)\overrightarrow{\mu}\left(0\right)\rho\left(0\right)\right]$$
(3.26)

The four correlation functions are distinct based on the number and order of their bra and ket interactions: $R_1^{(3)} = ket - ket - ket$, $R_2^{(3)} = bra - ket - bra$, $R_3^{(3)} = bra - bra - ket$, and $R_4^{(3)} = ket - bra - bra$. The Feynman diagrams of the four unique correlation functions for a two-level system are illustrated in figure 3.3.

3.2.1 Wavevector Matching

In general, third-order nonlinear spectroscopies involve multiple interacting beams. When the interacting beams have different geometries, their momenta (or wavevectors) must be conserved during the nonlinear process. Similarly, when the interacting pulses have finite spectral bandwidths, their energies must also be conserved. In third-order experiments, these



Figure 3.4: Wavevector matching. (a) BOXCARS beam geometry. (b) Wavevector mismatch dependent nonlinear signal intensity, where $L = 1 \ \mu m$ and $\lambda = 800 \ nm$.

conservation laws are dependent on the nonlinear process and take the general form:

$$\overrightarrow{k}_{4} = \pm \overrightarrow{k}_{1} \pm \overrightarrow{k}_{2} \pm \overrightarrow{k}_{3}$$
(3.27)

$$\omega_4 = \pm \omega_1 \pm \omega_2 \pm \omega_3 \tag{3.28}$$

where \vec{k}_i and ω_i are the wavevector and frequency components of beam *i*, respectively. Figure 3.4a illustrates a typical BOXCARS geometry. Energy conservation is fairly intuitive, while momentum conservation in nonlinear spectroscopy can be less so. However, it has been shown that the intensity of the nonlinear experiment strongly depends on momentum conservation:

$$I_4 \propto sinc^2 \left(\frac{\Delta kL}{2\pi}\right) \tag{3.29}$$

where $\Delta k = k_4 \mp k_1 \mp k_2 \mp k_3$ is the wavevector mismatch or deviation from perfect momentum conservation and L is the length of the nonlinear medium [107]. The dependence of the nonlinear signal intensity on the wavevector mismatch is illustrated in figure 3.4b.

An important consequence of momentum conservation is that signals of different nonlinear processes are radiated in different spatial directions depending on the number and type of interactions of each incident beam, called wavevector matching. Thus by controlling the geometry of the interacting beams, different third-order signals can be isolated by detection in a specific spatial direction. Additionally, the relative timing of the pulses composing the interacting beams can be used to detect different four-wave mixing signals in the same direction—as in the case of transient grating and two-dimensional spectroscopies.

3.3 Nonlinear Spectroscopic Techniques

We will now discuss the most common third-order, nonlinear spectroscopies. We start with the conventional and intuitive pump-probe spectroscopy. From there, we will extend our discussion to transient grating spectroscopy. This discussion will naturally lead to a description of the four types of third-order, two-dimensional (2D) spectroscopy. Finally, we will end by generalizing our discussion of 2D spectroscopy to higher-orders.

3.3.1 Pump-Probe Spectroscopy

In the classical picture, pump-probe spectroscopy uses a strong pump pulse to excite a material and after some time delay a second weak probe pulse interrogates the excited state. The change in the transmission of the probe pulse (i.e. linear absorption) as a function of time delay is used to interrogate the temporal dynamics of the material. Such an experiment is graphically illustrated in figure 3.5.

In the quantum mechanical picture, two pulses are incident on a material with the first pulse contributing two field interactions and the second pulse contributing one field interaction. The first "pump" pulse (wavevector \vec{k}_1) creates an excited state population $|e\rangle \langle e|$, initiating population decay during the delay time T between the first and second pulses. Additionally, if the excitation is coherent and the bandwidth sufficiently broad to cover multiple excited states, a coherence can be created between two excited states $|e'\rangle \langle e|$, initiating oscillations at the difference frequency $\omega_{e'e}$ and dephasing of the system during the delay time T. At some delay time T later, the second "probe" pulse (\vec{k}_2) projects the system onto a coherence $|g\rangle \langle e|$ and creates a third-order polarization that radiates the pump-probe signal $(\vec{k}_{PP} = \pm \vec{k}_1 \mp \vec{k}_1 + \vec{k}_2 = \vec{k}_2)$ in the direction of the probe pulse–called self-heterodyne



Figure 3.5: Pump-probe experiment. (a) Laser beam geometry. (b) Experimental diagram. (c) Pulse timing diagram. (d) Feynman diagram. (e) Energy level diagram. (d) Simulated emission energy-integrated pump-probe signal.

detection. In the frequency domain, the electromagnetic field amplitude of the pump-probe signal takes the form:

$$E_{Signal}^{(3)}(T,\omega_t) = i\chi_{PP}^{(3)}(T,\omega_t) E_1(T,\omega_t) E_1^*(T,\omega_t) E_2(T,\omega_t)$$
(3.30)

with contributions from all the correlation functions illustrated in figure 3.3 with the first time delay $\tau = 0$.

Similar to linear absorption, the pump-probe signal is locked $\frac{\pi}{2}$ out of phase from the probe pulse. Interference of the signal and probe pulses generates a change in the transmission of the probe pulse, which is measured in a spectrometer as a function of the delay time. The intensity of the transmitted electromagnetic field of the total pump-probe signal

 $I_{PP}^{(3)}(T,\omega_t)$ is measured by a spectrometer. The measured spectral output takes the form:

$$I_{PP}^{(3)}(T,\omega_t) = \left| E_2(T,\omega_t) + E_{Signal}^{(3)}(T,\omega_t) \right|^2$$

= $I_2(T,\omega_t) - 2\chi_{PP:\Im}^{(3)}(T,\omega_t) I_1(T,\omega_t) I_2(T,\omega_t)$ (3.31)

under the assumption $E_1 \gg E_2 \gg \chi_{PP}^{(3)}$.

Pump-probe measurements are designed to interrogate the effect of the pump pulse on the transmittance of the probe pulse, thus the intensity change in the probe pulse $\Delta I_{PP}^{(3)}(T,\omega_t)$ is used to determine the temporal dynamics in the material:

$$\Delta I_{PP}^{(3)}(T,\omega_t) = I_{PP}^{(3)}(T,\omega_t) - I_2(T,\omega_t)$$

= $-2\chi_{PP:\Im}^{(3)}(T,\omega_t) I_1(T,\omega_t) I_2(T,\omega_t)$ (3.32)

The pump-probe signal intensity in equation 3.31 and the change in probe transmission in equation 3.32 are directly analogous to the case of linear absorption in equation 3.5 under the influence of a pump pulse incident at a time delay T prior to the probe pulse. The change in the transmission of the probe pulse occurs due to the imaginary (i.e. absorptive) component of the third-order susceptibility $\chi^{(3)}_{PP;\Im}$ because the electromagnetic field of the pump-probe signal is locked $\frac{\pi}{2}$ out of phase from the probe pulse.

3.3.2 Transient Grating Spectroscopy

In the classical picture, transient grating spectroscopy is an extension of pump-probe spectroscopy in which the pump and probe pulses are each split into two beams. First, the two time-coincident pump pulses are crossed at an angle θ in the material to create an excited state (transient) grating due to the interference of the two beams, illustrated in figure 3.6. The period of the transient grating is given by



Figure 3.6: Transient grating spatial periodicity. Transient grating spatial periodicity $L = \lambda/2 \sin [\theta/2]$, where λ is the center wavelength of the two incident beams and θ is the angle between them.

$$L = \frac{\lambda}{2\sin\left(\theta/2\right)} \tag{3.33}$$

where λ is the central wavelength of the pump pulses. At some delay time T later, a third probe pulse diffracts off the excited state grating into the direction of a fourth heterodyning pulse. The change in the diffraction of the probe pulse as a function of time delay is used to measure the temporal dynamics of the material. The components of a transient grating experiment are illustrated in figure 3.7. The spatially periodicity of the transient grating can also be used to measure transport [108, 109, 110].

In the quantum mechanical picture, three pulses are incident on a material with each contributing a single field interaction. The first pulse (wavevector $\vec{k_1}$) creates a coherence between the ground state and an excited electronic state $|g\rangle \langle e|$. The second pulse ($\vec{k_2}$) is time-coincident with the first pulse ($\tau = 0$) and creates an excited state population $|e\rangle \langle e|$, initiating population decay of the system during the delay time T between the second and third pulses. Additionally, if the excitation is coherent and the bandwidth of the pump pulses is sufficiently broad to cover multiple excited states, a coherence can be created



Figure 3.7: Transient grating experiment. (a) Laser beam geometry. (b) Experimental diagram. (c) Pulse timing diagram. (d) Feynman diagram. (e) Energy level diagram. (f) Simulated transient grating signal.

between two excited states $|e'\rangle \langle e|$, initiating oscillations at the difference frequency $\omega_{e'e}$ and dephasing of the system during the delay time T. At some delay time T later, the third pulse (\vec{k}_3) projects the system onto a coherence $|g\rangle \langle e|$ and creates a third-order polarization that radiates the transient grating signal $(\vec{k}_{TG} = \pm \vec{k}_1 \mp \vec{k}_2 + \vec{k}_3)$. The emitted signal is heterodyne-detected with a fourth pulse $(\vec{k}_4 = \vec{k}_{TG})$. In the frequency domain, the electromagnetic field amplitude of the transient grating signal takes the form:

$$E_{Signal}^{(3)}(T,\omega_t) = i\chi_{TG}^{(3)}(T,\omega_t) E_1(T,\omega_t) E_2^*(T,\omega_t) E_3(T,\omega_t)$$
(3.34)

with contributions from all the correlation functions illustrated in figure 3.3 with the first time delay $\tau = 0$.

Unlike pump-probe spectroscopy, the transient grating signal is not self-heterodyned. The phase between the signal and the fourth heterodyning pulse $\Delta \phi_{HD}$ can be experimentally controlled. Thus, the electromagnetic field intensity of the total transient grating signal

 $I_{TG}^{(3)}(T,\omega_t)$ takes the form:

$$I_{TG}^{(3)}(T,\omega_{t}) = \left| E_{HD}(T,\omega_{t}) \exp\left(i\Delta\phi_{HD}\right) + E_{Signal}^{(3)}(T,\omega_{t}) \right|^{2}$$

= $I_{HD}(T,\omega_{t}) - 2 \left[\chi_{TG:\Re}^{(3)}(T,\omega_{t}) \sin\left(\Delta\phi_{HD}\right) + \chi_{TG:\Im}^{(3)}(T,\omega_{t}) \cos\left(\Delta\phi_{HD}\right) \right]$
 $\times E_{1}(T,\omega_{t}) E_{2}^{*}(T,\omega_{t}) E_{3}(T,\omega_{t}) E_{HD}^{*}(T,\omega_{t})$ (3.35)

under the assumption $E_i \gg \chi_{TG}^{(3)}$. When $\Delta \phi_{HD} = \pi \left(n + \frac{1}{2}\right)$ where *n* is an integer, the electromagnetic field intensity of the transient grating signal is analogous to the pump-probe case (equation 3.31):

$$I_{TG}^{(3)}(T,\omega_t) = I_{HD}(T,\omega_t) - 2\chi_{TG:\Im}^{(3)}(T,\omega_t) E_1(T,\omega_t) E_2^*(T,\omega_t) E_3(T,\omega_t) E_{HD}^*(T,\omega_t)$$
(3.36)

and the dynamics related to the imaginary (i.e. absorptive) component of the third-order susceptibility $\chi^{(3)}_{TG:\Im}$ are measured. When $\Delta \phi_{HD} = n\pi$ where *n* is an integer, the transient grating signal reveals the dynamics related to the real (i.e. dispersive) component of the third-order susceptibility $\chi^{(3)}_{TG:\Re}$:

$$I_{TG}^{(3)}(T,\omega_t) = I_{HD}(T,\omega_t) - 2\chi_{TG:\Re}^{(3)}(T,\omega_t) E_1(T,\omega_t) E_2^*(T,\omega_t) E_3(T,\omega_t) E_{HD}^*(T,\omega_t)$$
(3.37)

In transient grating spectroscopy, separation of the pump and probe pulses each into two beams creates more experimental input parameters that allow both the absorptive and dispersive components of the third-order susceptibility to be measured [111]. Additionally, the spatial periodicity of the transient grating allows measurement of transport properties. Thus, we can see that increasing the number of experimental input parameters increases the number and type of measurable spectroscopic signals.



Figure 3.8: Two-dimensional rephasing experiment. (a) Laser beam geometry. (b) Pulse timing diagram. (c) Feynman diagram. (d) Energy level diagram. (e) Time-dependent phase evolution.

3.3.3 Two-Dimensional Rephasing (Photon Echo) Spectroscopy

In our discussion of 2D spectroscopy [39, 40], we will start with the quantum mechanical picture because the classical picture is less intuitive than those for pump-probe and transient grating spectroscopies. Before we begin, we note that 2D spectroscopy is analogous to transient grating spectroscopy in that it requires the interaction of three pulses in a material and a fourth heterodyning pulse to detect the signal. However, we increase the number of experimental input parameters by changing the time delay between the first and second pulses to measure new spectroscopic signals.

In a 2D rephasing measurement (illustrated in figure 3.8), three pulses are incident on a material with each contributing a single field interaction. The first pulse (wavevector $-\vec{k_1}$) creates a coherence between the ground and an excited electronic state $|g\rangle \langle e|$, initiating oscillations at the central frequency ω_{ge} and dephasing of the system during the excitation time τ between the first and second pulses. Subsequently, the second pulse $(\vec{k_2})$ creates an excited state population $|e\rangle \langle e|$ that decays during the population time T between the second and third pulses. Additionally, if the excitation is coherent and the bandwidth of

the first two pulses is sufficiently broad to cover multiple excited states, a coherence can be created between two excited states $|e'\rangle \langle e|$, initiating oscillations at the difference frequency $\omega_{e'e}$ and dephasing of the system during the delay time *T*. Finally, the third pulse (\vec{k}_3) creates a coherence $|e\rangle \langle g|$, conjugate to that of the excitation time, initiating oscillations and rephasing of the inhomogeneous frequency components of the system around the central frequency ω_{ge} during the emission time *t* after the third pulse. At time $t = \tau$, the system is rephased and the constructive superposition of the frequency components is maximized and emits a photon echo ($\vec{k}_R = -\vec{k}_1 + \vec{k}_2 + \vec{k}_3$). The emitted signal is heterodyne-detected with a fourth pulse ($\vec{k}_4 = \vec{k}_{TG}$) by spectral interferometry to recover the full electric field. In the frequency domain, the electromagnetic field amplitude of the rephasing signal takes the form:

$$E_{Signal}^{(3)}(\tau, T, \omega_t) = i\chi_R^{(3)}(\tau, T, \omega_t) E_1^*(\tau, T, \omega_t) E_2(\tau, T, \omega_t) E_3(\tau, T, \omega_t)$$
(3.38)

with contributions from the $R_2^{(3)}$ and $R_3^{(3)}$ correlation functions illustrated in figure 3.3.

The total rephasing signal intensity is measured by a spectrometer as a function of the excitation time, $I_R^{(3)}(\tau, T, \omega_t)$, and subsequently Fourier transformed to produce the 2D rephasing spectrum, $I_R^{(3)}(\omega_{\tau}, T, \omega_t)$, illustrated in figure 3.9. The electromagnetic field intensity of the rephasing signal is analogous to the transient grating case (equation 3.35) with the addition of the excitation time delay τ :

$$I_{R}^{(3)}(\omega_{t},T,\omega_{t}) = I_{HD}(\omega_{t},T,\omega_{t}) - 2\left[\chi_{R:\Re}^{(3)}(\omega_{t},T,\omega_{t})\sin(\phi_{HD}) + \chi_{R:\Im}^{(3)}(\omega_{t},T,\omega_{t})\cos(\phi_{HD})\right] \times E_{1}^{*}(\omega_{t},T,\omega_{t})E_{2}(\omega_{t},T,\omega_{t})E_{3}(\omega_{t},T,\omega_{t})E_{HD}^{*}(\omega_{t},T,\omega_{t})$$
(3.39)

The rephasing of the inhomogeneous components of the system during the emission time t allows the homogeneous linewidth of the material resonance to be measured by 2D rephasing spectroscopy. In the 2D rephasing spectrum (figure 3.9), the diagonal and anti-diagonal linewidths correspond to the inhomogeneous (Γ) and homogeneous (Δ) linewidths, respec-



Figure 3.9: Two-dimensional rephasing spectrum. (a) Simulated rephasing spectrum reveals the homogenous Γ and inhomogeneous Δ linewidths. (b) Diagonal and off-diagonal Feynman diagrams.

tively. The ability of 2D rephasing experiments to measure homogeneous linewidths can be understood in the classical picture.

The first 2D rephasing spectroscopy experiments were performed by nuclear magnetic resonance (NMR) using electromagnetic waves in the MHz to GHz range to interrogate nuclear magnetic spin dynamics, illustrated in figure 3.10 [112]. In this experiment, a strong static magnetic field \vec{B} is used to align the magnetic nuclear spins of atoms or molecules. An electromagnetic pulse perpendicular to static magnetic field aligns some of the nuclear spins in the plane. Once the electromagnetic pulse is gone, the nuclear spins in the plane begin to precess due to the torque $\vec{\tau}$ exerted on the magnetic dipole moment $\vec{\mu}_B$ by the perpendicular magnetic field, $\vec{\tau} = \vec{\mu}_B \times \vec{B}$. The angular frequency of precession can be written as $\omega_B = -\gamma B$, where $\gamma = \frac{q}{2m}$ is the gyromagnetic ratio with q the charge of the particle, m the mass of the particle, and B the amplitude of the static magnetic field. Because the angular frequency of precession is dependent on the charge and mass of the particle, the nuclear spins spread out according to their gyromagnetic ratios during the excitation time τ . After some variable excitation time τ later, a second electromagnetic pulse antiparallel



Figure 3.10: Two-dimensional spin echo in NMR. (1) Nuclear magnetic spins are aligned by a strong static field B_z . (2) A perpendicular magnetic pulse B_x aligns a subset of nuclear spins into the xy plane. (3) Nuclear spins in the xy plane precess at different frequencies according to their mass and charge due to the torque exerted by the static magnetic field B_z . (4) At some delay time τ later, a second magnetic pulse $-B_x$ flips the nuclear spins in the xy plane. (5) The nuclear spins begin to converge in the xy plane. (6) At 2τ , the nuclear spins realign and their macroscopic oscillating polarization radiates a spin echo.

to the first pulse flips the nuclear spins in the plane. Once the electromagnetic field is gone, the nuclear spins continue to precess in the same direction. However, since the nuclear spins have been flipped in the plane, the nuclear spins begin to converge during the emission time t. When $t = \tau$, the nuclear spins are realigned in the plane and their macroscopic oscillating polarization emits a photon echo.

2D rephasing experiments of excited electronic states are analogous to the case of nuclear magnetic spins in NMR. The first pulse $(-\vec{k}_1)$ creates multiple excited state coherences that initiate oscillations at their resonance frequencies. During the excitation time τ , the system dephases due to differences in the resonance frequencies. After some variable excitation time τ later, the second (\vec{k}_2) and third (\vec{k}_3) pulses (T = 0) "flip" the system by creating the coherences conjugate to those during the excitation time. During the excitation time t, the system rephases. When $t = \tau$, the rephasing of the excited states is maximized and their macroscopic oscillating polarization emits a photon echo.

3.3.4 Two-Dimensional Non-Rephasing Spectroscopy

In a 2D non-rephasing measurement (illustrated in figure 3.11), three pulses are incident on a material with each contributing a single field interaction. The first pulse (wavevector \vec{k}_1) creates a coherence between the ground and an excited electronic state $|e\rangle \langle g|$, initiating oscillations at the central frequency ω_{eg} and dephasing of the system during the excitation time τ between the first and second pulses. Subsequently, the second pulse $(-\vec{k}_2)$ creates an excited state population $|e\rangle \langle e|$ that decays during the population time T between the second and third pulses. Additionally, if the excitation is coherent and the bandwidth of the first two pulses is sufficiently broad to cover multiple excited states, a coherence can be created between two excited states $|e\rangle \langle e'|$, initiating oscillations at the difference frequency $\omega_{ee'}$ and dephasing of the system during the delay time T. Finally, the third pulse (\vec{k}_3) creates a coherence $|e\rangle \langle g|$, the same as that of the excitation time, initiating oscillations and further dephasing of the system during the emission time t after the third pulse. During the emission



Figure 3.11: Two-dimensional non-rephasing experiment. (a) Laser beam geometry. (b) Pulse timing diagram. (c) Feynman diagram. (d) Energy level diagram. (e) Time-dependent phase evolution.

time t, the third-order polarization generated by the three incident pulses radiates to produce the non-rephasing signal $(\vec{k}_{NR} = \vec{k}_1 - \vec{k}_2 + \vec{k}_3)$. The emitted signal is heterodyne-detected with a fourth pulse $(\vec{k}_4 = \vec{k}_{NR})$ by spectral interferometry to recover the full electric field. In the frequency domain, the electromagnetic field amplitude of the non-rephasing signal takes the form:

$$E_{Signal}^{(3)}(\tau, T, \omega_t) = i\chi_{NR}^{(3)}(\tau, T, \omega_t) E_1(\tau, T, \omega_t) E_2^*(\tau, T, \omega_t) E_3(\tau, T, \omega_t)$$
(3.40)

with contributions from the $R_1^{(3)}$ and $R_4^{(3)}$ correlation functions illustrated in figure 3.3.

The emitted signal is measured by a spectrometer as a function of the excitation time, $I_{NR}^{(3)}(\tau, T, \omega_t)$, and subsequently Fourier transformed to produce the 2D non-rephasing spectrum, $I_R^{(3)}(\omega_{\tau}, T, \omega_t)$, illustrated in figure 3.12. The electromagnetic field intensity of the non-rephasing signal is analogous to the rephasing signal (equation 3.39):

$$I_{NR}^{(3)}(\omega_t, T, \omega_t) = I_{HD}(\omega_t, T, \omega_t) - 2 \left[\chi_{NR:\Re}^{(3)}(\omega_t, T, \omega_t) \sin(\phi_{HD}) + \chi_{NR:\Im}^{(3)}(\omega_t, T, \omega_t) \cos(\phi_{HD}) \right] \\ \times E_1(\omega_t, T, \omega_t) E_2^*(\omega_t, T, \omega_t) E_3(\omega_t, T, \omega_t) E_{HD}^*(\omega_t, T, \omega_t)$$
(3.41)



Figure 3.12: Two-dimensional non-rephasing spectrum. (a) Simulated non-rephasing spectrum. (b) Diagonal and off-diagonal Feynman diagrams.

Interpretation of 2D rephasing and non-rephasing spectra (figures 3.9 and 3.12, respectively) is very similar. Diagonal peaks correspond to spectral features in the linear absorption spectrum, while off-diagonal peaks demonstrate coupling between spectral features but do not reveal how features are coupled-often through a common ground state. By measuring multiple 2D spectra as a function of the population time T, population transfer and coherent coupling between excited states can be measured. Coherent coupling manifests itself differently in rephasing and non-rephasing spectra. In the rephasing spectrum, off-diagonal peaks oscillate at the difference frequency of the coupled excited states as a function of the population time T. In the non-rephasing spectrum, diagonal peaks oscillate. Collecting 2D spectra with sufficient population time resolution, a 3D spectrum can be generated that fully characterizes the third-order spectroscopic response of the interrogated excited states [63].

3.3.5 Two-Dimensional Correlation Spectroscopy

The astute reader will have noticed that neither the rephasing nor the non-rephasing 2D spectra had contributions from all the correlation functions for the third-order response function. The rephasing spectrum corresponds to the $R_1^{(3)}$ and $R_4^{(3)}$ correlation functions and



Figure 3.13: Two-dimensional correlation spectrum. (a) Simulated correlation spectrum. (b) Absorptive and dispersive components and excitation energy-integrated projections.

the non-rephasing spectrum to the $R_2^{(3)}$ and $R_3^{(3)}$ correlation functions in figure 3.3. Because neither comprises the full set of correlation functions, the 2D spectra have a phase twist and result in imaginary and real components that no longer correspond to the absorptive or dispersive components, respectively, but rather to linear combinations [113].

To separate the absorptive and dispersive components of the 2D signal, the equally weighted rephasing and non-rephasing contributions must be summed. The 2D correlation spectrum contains all the correlation functions of the third-order response, eliminating the phase twist and resulting in a spectrum for which the absorptive and dispersive components are separable. Figure 3.13 shows a 2D correlation spectrum and the separable absorptive and dispersive components of the signal.

3.3.6 Two-Dimensional Two-Quantum Spectroscopy

In a two-quantum measurement (illustrated in figure 3.14) [114, 61], three pulses are incident on a material with each contributing a single field interaction. The first pulse (wavevector \vec{k}_1) creates a coherence between the ground state and an excited electronic state $|g\rangle \langle e|$. The second pulse (\vec{k}_2) is time-coincident with the first pulse ($\tau = 0$) and creates a two-quantum



Figure 3.14: Two-dimensional two-quantum experiment. (a) Laser beam geometry. (b) Pulse timing diagram. (c) Feynman diagram. (d) Energy level diagram. (e) Time-dependent phase evolution.

coherence between the ground state and a doubly excited electronic state $|g\rangle \langle b|$, initiating oscillations at the two-quantum frequency ω_{gb} . At some two-quantum time T_{2Q} later, the third pulse $(-\vec{k}_3)$ projects the system onto a one-quantum coherence $|g\rangle \langle e|$ and creates a third-order polarization that radiates the two-quantum signal $(\vec{k}_{2Q} = \vec{k}_1 + \vec{k}_2 - \vec{k}_3)$. The emitted signal is heterodyne-detected with a fourth pulse $(\vec{k}_4 = \vec{k}_{2Q})$ by spectral interferometry to recover the full electric field. In the frequency domain, the electromagnetic field amplitude of the two-quantum signal takes the form:

$$E_{Signal}^{(3)}\left(0, T_{2Q}, \omega_{t}\right) = i\chi_{2Q}^{(3)}\left(0, T_{2Q}, \omega_{t}\right) E_{1}\left(0, T_{2Q}, \omega_{t}\right) E_{2}\left(0, T_{2Q}, \omega_{t}\right) E_{3}^{*}\left(0, T_{2Q}, \omega_{t}\right)$$
(3.42)

with the related correlation functions illustrated in figure 3.15.

The emitted signal is measured by a spectrometer as a function of the two-quantum time, $I_{2Q}^{(3)}(0, T_{2Q}, \omega_t)$, and subsequently Fourier transformed to produce the two-quantum spectrum, $I_{2Q}^{(3)}(0, \omega_{2Q}, \omega_t)$, illustrated in figure 3.16. The electromagnetic field intensity of the two-quantum signal is analogous to the rephasing and non-rephasing signals (equations



Figure 3.15: Two-quantum Feynman diagrams. (a) Unique correlation functions of the third-order, two-quantum response function $R_{2Q}^{(3)}(0, T_{2Q}, t)$. (b) Three-level system energy level energy diagram, including the ground g, single-exciton e, and biexciton b manifolds.



Figure 3.16: Two-dimensional two-quantum spectrum. (a) Simulated two-quantum spectrum. (b) Diagonal and off-diagonal Feynman diagrams.

3.39 and 3.41, respectively):

$$I_{2Q}^{(3)}(0, T_{2Q}, \omega_t) = I_{HD}(0, T_{2Q}, \omega_t)$$

$$-2 \left[\chi_{2Q:\Re}^{(3)}(0, T_{2Q}, \omega_t) \sin(\phi_{HD}) + \chi_{2Q:\Im}^{(3)}(0, T_{2Q}, \omega_t) \cos(\phi_{HD}) \right]$$

$$\times E_1(0, T_{2Q}, \omega_t) E_2(0, T_{2Q}, \omega_t) E_3^*(0, T_{2Q}, \omega_t) E_{HD}^*(0, T_{2Q}, \omega_t)$$
(3.43)

In the two-quantum spectrum, the excitation energy (called the two-quantum energy) is twice the magnitude of the rephasing and non-rephasing spectra, while the emission energy remains the same. The spectral features of two-particle correlations lie on the y = 2x line rather than the y = x line as in the case of one-particle correlations in the rephasing and non-rephasing spectra. Spectral features on the y = 2x line correspond with two correlated particles of the same type (AA correlations), while off-diagonal features correspond with two correlated particles of different types (mixed AB correlations). Deviation of the two-quantum energy of the two-particle correlation from the sum of their constituent particle energies reveals the two-particle binding energy (or anharmonicity), $\Delta E_{2P} = (E_{p1} + E_{p2}) - E_{2P}$. Direct observation of two-particle correlations well separated from strong one-particle signals using 2D two-quantum spectroscopy (unlike one-quantum spectroscopies) is a fundamental experimental advantage that has revealed substantial information about the types, energetic properties, and temporal dynamics of correlated systems.

3.3.7 Higher-Order Multi-Quantum Spectroscopy

Correlations of more than two excitations can be measured using higher-order spectroscopies. To date, three correlated excitons in GaAs quantum wells and four correlated exciton-polaritons in GaAs quantum wells embedded in a microcavity have been measured using fifth- and seventh-order spectroscopies [62, 100]. Qualitatively, these experiments are analogous to third-order two-quantum spectroscopy. A correlation of N particles can be measured with an experiment that creates an N-quantum coherence between the ground and N-particle state $|N\rangle \langle g|$ by N field interactions. The dynamics of the N-particle correlation are measured by the oscillations at the N-quantum frequency ω_{gN} and the dephasing of the system. At some N-quantum time T_{NQ} later, N-1 field interactions project the system onto a one-quantum coherence $|g\rangle \langle e|$ and creates a (2N-1)-order polarization that radiates the N-quantum signal and reveals the N-particle correlation.

3.4 Spectral Lineshapes

Previously, the concept of the time-dependent density matrix was introduced via the von Neumann relation (equation 3.12) to describe the system of an ensemble of quantum states. The von Neumann relation is now recast in its derivative form to determine the kinetics of the density matrix, $\rho = \sum_{i,j} |i\rangle \langle j|$.

$$\frac{\partial \rho}{\partial t} = \frac{i}{\hbar} \left[\hat{H}, \rho \right] \tag{3.44}$$

where is \hat{H} the Hamiltonian. A Hamiltonian describing the state of a quantum system under electromagnetic excitation can be written as the sum of a system Hamiltonian \hat{H}_0 , relaxation Hamiltonian $\hat{H}_R = -\Gamma \times \rho$, and time-dependent excitation potential $\hat{V}(t) = -\overrightarrow{\mu} \times \overrightarrow{E}(t)$.

$$\hat{H} = \hat{H}_0 + \hat{H}_R + \hat{V}(t)$$

$$= \begin{bmatrix} \hbar \omega_g & 0 \\ 0 & \hbar \omega_e \end{bmatrix} - \begin{bmatrix} 0 & -\Gamma_{ge} \\ -\Gamma_{eg} & -\Gamma_{ee} \end{bmatrix} \begin{bmatrix} \rho_{gg} & \rho_{ge} \\ \rho_{eg} & \rho_{ee} \end{bmatrix} - \begin{bmatrix} 0 & \mu_{ge} \\ \mu_{eg} & 0 \end{bmatrix} E(t) \quad (3.45)$$

where $\hbar\omega_n$ are the state energies, Γ_{nn} and Γ_{nm} the population and coherence lifetimes, ρ_{nm} the density matrix elements, μ_{nm} the transition dipole moments, and E(t) the exciting electromagnetic field. Substituting the Hamiltonian (equation 3.45) into the von Neumann relation (equation 3.44), the time-derivative of the density matrix for the two-level system takes the form:

$$\frac{\partial \rho}{\partial t} = \frac{i}{\hbar} \begin{bmatrix} \Gamma_{ee}\rho_{ee} + (\mu_{eg}\rho_{ge} - \mu_{ge}\rho_{eg}) E(t) & (\hbar\omega_{ge} + i\Gamma_{ge}) \rho_{ge} + \mu_{ge} (\rho_{gg} - \rho_{ee}) E(t) \\ (\hbar\omega_{eg} + i\Gamma_{eg}) \rho_{eg} + \mu_{eg} (\rho_{ee} - \rho_{gg}) E(t) & -\Gamma_{ee}\rho_{ee} + (\mu_{ge}\rho_{eg} - \mu_{eg}\rho_{ge}) E(t) \end{bmatrix}$$
(3.46)

where ω_{nm} is the frequency difference between the states *n* and *m*. The $(\hbar\omega_{nm} + i\Gamma_{nm})\rho_{nm}$ terms determine the frequency and spectral lineshape of the excitation. The $\pm\Gamma_{nn}\rho_{nn}$ terms represent population relaxation from the excited state to the ground state. The transition dipole moment terms μ_{nm} represent the optical excitation from the ground state into and out of the excited state.

Integrating equation 3.46 produces solutions to the off-diagonal density matrix elements of the form $\rho_{ge}(t) = \exp\left[\left(i\omega_{ge} - \frac{\Gamma_{ge}}{\hbar}\right)t\right]$, whose Fourier transformations yield Lorentzian lineshapes of width $\gamma_{ge} = \frac{\Gamma_{ge}}{\hbar}$ at the central frequency ω_{ge} :

$$\rho_{ge}\left(\omega\right) = \frac{1}{\pi\gamma_{ge}} \frac{\gamma_{ge}^{2}}{\left(\left(\omega - \omega_{ge}\right)^{2} + \gamma_{ge}^{2}\right)}$$
(3.47)

3.4.1 Local Field Effects

Under strong field excitation, local fields are generated due to the dipole moment of electronhole pairs (i.e. excitons) [115, 116]. Thus, the total electromagnetic field acting on the system takes the form: $E(t) = E_0(t) + E_L(t)$. Exciton states are destabilized due to the presence of the local field $E_L(t)$ [117, 118, 119]. However, local fields also induce slower system decoherence as excitations continue to be driven after the applied electromagnetic field is gone [120, 121]. These effects can be introduced phenomenologically by adding densitydependent frequency and coherence lifetimes terms into the equation 3.46:

$$\frac{\partial \rho_{ge}}{\partial t} = \left[i\left(\omega_{ge} + N\omega^{EIS}\right) - \left(\gamma_{ge} + N\gamma^{EID}\right)\right]\rho_{ge} \tag{3.48}$$

where N is the excitation density, ω^{EIS} describes excitation-induced frequency shift, and γ^{EID} describes excitation-induced dephasing.
Chapter 4

Experimental Setup

In chapter 4, we will first discuss the ultrafast laser system and non-collinear optical parametric amplifier used for the experiments in this thesis [122]. From there, we will describe our spatial-temporal based two-dimensional spectrometer, including the theory of temporal pulse shaping and the calibrations and other technical considerations required for an ultrafast experiment [114, 123]. Finally, we will discuss the detection methods used for nonlinear spectroscopy: rotating wave detection, spectral interferometry, phase cycling, and the phase stability of our two-dimensional spectrometer [124, 125, 126].

4.1 Ultrafast Laser System

The experiments described in this thesis were all performed using a Coherent Libra titanium sapphire (Ti:Sapph) regenerative amplifier system [127]. The output of the Libra was a 3.0 W, 10 kHz pulse train of approximately 100 fs pulses centered at 800 nm with a full width at half maximum of 10 nm.



Figure 4.1: Experimental non-collinear optical parametric amplifier. 1 W of 100 fs, 800 nm pulses are incident on a 95:5 beamsplitter (BS). 95% is transmitted through the BS and onto a delay stage. The 0.95 W beam passes through a waveplate/polarizer pair (WP/P) for attenuation and then into a Beta Barium Borate crystal (BBO) for Type I second harmonic generation. The 400 nm pump beam is then focused by a 40 cm lens (L1) onto the mixing BBO for optical parametric amplification. The 0.05 W beam reflected off the BS is attenuated by an iris (I) and focused by a 3 cm lens (L2) into a sapphire plate (SP) for white light continuum generation. The white light seed beam is then single lens imaged by a 10 cm lens (L3) onto the mixing BBO for optical parametric amplified visible beam is filtered out by a shortpass filter (SPF). The spatial mode of the beam is subsequently fixed by focusing through a 200 μ m pinhole and the pulse compressed and any spatial chirp corrected by a prism compressor.

4.2 Non-Collinear Optical Parametric Amplifier

To perform experiments throughout the visible region of the electromagnetic spectrum, the 800 nm output of the regenerative amplifier must be upconverted to the region of 500 - 700 nm.

Upconversion is accomplished by optical parametric amplification (OPA) using a noncollinear optical parametric amplifier (NOPA) [122]. First, a large fraction of the input beam is upconverted by second harmonic generation (SHG). In the experimental setup, approximately 0.95 W of the 800 nm beam is transmitted through a beamsplitter, traverses a delay stage, is attenuated by a waveplate/polarizer pair, and propagates through a Beta Barium Borate (BBO) crystal optimized for Type I SHG to generate 400 nm light. Type I SHG is a second-order nonlinear process in which two photons at the fundamental frequency with the same polarization sum to produce one photon at twice the fundamental frequency, $\omega_{SHG} = 2\omega_F$, with a polarization orthogonal to the fundamental [107]. The process of SHG can mathematically written in the frequency domain as:

$$E(\omega_{SHG}) \sim i P_{SHG}^{(2)}(2\omega_F) = i \chi_{SHG}^{(2)} E(\omega_F) E(\omega_F)$$
(4.1)

where $E(\omega_{SHG})$ is the spectral amplitude of the second harmonic, $P_{SHG}^{(2)}(2\omega_F)$ is the secondorder nonlinear polarization generated at the second harmonic frequency in the doubling BBO, $\chi_{SHG}^{(2)}$ is the second-order nonlinear susceptibility of the doubling BBO, and $E(\omega_F)$ is the spectral amplitude of the fundamental. Second-order nonlinear processes are only possible in non-centrosymmetric materials, such as BBO, as the crystal symmetry must allow for finite values of the second-order susceptibility.

Once the 800 nm fundamental has been upconverted to 400 nm, it must be downconverted from the ultraviolet into the visible by OPA. OPA is a second-order nonlinear process in which a strong pump beam (P) is downconverted using a weak seed beam (S) by difference frequency generation, $\omega_P = \omega_S + \omega_I$, and results in the amplification of the frequencies of the seed beam. In the nonlinear process, energy and momentum must be conserved by producing a third idler beam (I). The process of OPA can mathematically written in the frequency domain as:

$$E(\omega_S) \sim i P_{OPA}^{(2)}(\omega_S) = i \chi_{OPA}^{(2)} E(\omega_P) E(\omega_P)$$
(4.2)

where $E(\omega_S)$ is the spectral amplitude of the amplified seed beam, $P_{OPA}^{(2)}(\omega_S)$ is the secondorder nonlinear polarization generated at the seed frequencies in the mixing BBO, and $E(\omega_P)$ is the spectral amplitude of the pump beam.

To generate the seed beam, approximately 0.05 W of the 800 nm beam is reflected off the

beam splitter, attenuated by an iris, and focused with a 3 cm lens into a 1 mm thick sapphire plate to produce a white light continuum by self-phase modulation (SPM). SPM is a secondorder nonlinear process in which a large electric field intensity creates an intensity-dependent refractive index:

$$n(\omega, I) = n_0(\omega) + n^{(2)}(\omega) I$$
(4.3)

where $n_0(\omega)$ is the linear refractive index of a material, $n^{(2)}(\omega)$ is the second-order nonlinear refractive index, and I is the electric field intensity. The intensity-dependent refractive index produced as the light propagates through the sapphire generates a phase shift in the pulse that broadens the pulse in the frequency domain and produces a supercontinuum covering the entire visible spectrum [107]. The supercontinuum is subsequently imaged with a single 10 cm lens onto the mixing BBO optimized for OPA.

The 400 nm pump beam is simultaneously focused by a 40 cm lens into the mixing BBO at an angle α relative to the supercontinuum. Because indices of refraction are frequencydependent, dispersion occurs when different frequencies are generated or propagate through a material, limiting the intensity and bandwidth of the amplified light. However, BBO is birefringent so the indices of refraction along different crystallographic axes can be used to phasematch large bandwidths. The angle α between the pump and seed beams is chosen such that the dispersion (i.e. group velocity mismatch) between the seed and idler are matched along the direction of amplification, allowing for broadband frequency generation. In BBO for broadband generation centered at 600 nm, α is approximately 3.7°.

After the mixing BBO, the pump and idler beams are rejected and the amplified beam is collimated using a 15 cm lens. The remaining 800 nm light in the amplified beam is eliminated using a shortpass filter. The spatial profile of the beam is cleaned up by focusing through a 200 µm pinhole and a prism compressor is used to compress the pulses and correct for any spatial chirp generated during OPA.

With the NOPA, bandwidths spanning 500 to 700 nm can be easily achieved. However, such large bandwidths are not suitable for the rest of our experimental setup. Thus, the



Figure 4.2: Non-collinear optical parametric amplifier output. (a) NOPA spectrum at 600 nm with a full width at half the maximum of 40 nm. (b) NOPA pulse compressed by a pair of prisms to 40 fs and measured using a second harmonic generation frequency-resolved optical gating (SHG FROG).

NOPA output bandwidth is generally limited to approximately 45 nm by introducing dispersion into the white light arm and then tuned to the desired region of the visible spectrum by the time delay between the 400 nm pump beam and the white light supercontinuum. Figure 4.2 shows a typical narrowband NOPA spectrum and the corresponding pulse in the time domain.

4.3 Spatio-Temporal Pulse Shaping

Once visible ultrafast pulses have been generated in the NOPA and compressed by a prism pair, the beam is routed into our two-dimensional (2D) spectrometer for multiple beam generation and temporal pulse shaping. Figure 4.3 illustrates the components and layout of the 2D spectrometer including a Hamamatsu X7550 2D liquid crystal spatial light modulator and a Princeton Instruments Action SP2300 spectrometer with a PIXIS 400 CCD



Figure 4.3: Two-dimensional spectrometer. Ultrafast laser pulses from the NOPA are focused by a lens L1 onto a phase mask (PM) to produce four beams in the BOXCARS geometry and subsequently collimated by lens L2. Lens L3 and L4 relay image the beams into the temporal pulse shaper, consisting of a grating (Gr), cylindrical lens (CL), and 2D spatial light modulator (SLM). The grating spectrally disperses the beams and the cylindrical lens focuses them onto vertically distinct regions of the 2D SLM, which controls the frequency-dependent phase of each beam. The phase-perturbed beams are then refocused by the cylindrical lens on to the grating and are compressed. Lens L4 focuses the beams onto the pick-off mirror M1 and lens L5 recollimates them. Finally, lens L6 focuses the beams onto the sample and the signal and local oscillator are detected by a spectrometer.



Figure 4.4: Beam pattern generation. (a) Linear three beam geometry using a 1D phase mask. (b) BOXCARS four beam geometry using a 2D phase mask.

camera.

4.3.1 Principles of Pulse Shaping

4.3.1.1 Beam Pattern Generation

To generate multiple beams in a specified geometry, a phase mask (or low dispersion, transmissive, diffractive optic) optimized for first-order diffraction is used. A linear set of three beams can be generated by a phase mask with a square grating pattern in one dimension, figure 4.4a; a set of fours beams in the BOXCARS geometry can be generated by a phase mask with two perpendicular square grating patterns, figure 4.4b. To generate the desired beam set, the fundamental beam is focused by a lens onto the phase mask causing diffraction of the beam into multiple orders. The diffracted beams are collimated by a second lens and the desired beams for a specific geometry are imaged using common path optics to the temporal pulse shaper and subsequently to the sample position. Imaging of the phase mask pattern onto the sample position has the advantage of automatic phase matching of the wavevectors for each beam (including the local oscillator) and an increase in the measured 2D signal due to an enhancement of the phase front overlaps of the beams at the sample position [128].

4.3.1.2 Temporal Pulse Shaping

From the phase mask, the beams are imaged using common path optics into a pulse shaper consisting of a plane-ruled reflectance grating, achromatic cylindrical lens, and a 2D spatial light modulator (SLM). In the pulse shaper, the beams are spectrally dispersed by the grating, which is blazed at the center wavelength of the ultrafast pulse and has a periodicity of approximately 1,000 groves/mm. The cylindrical lens subsequently stops the divergence of the spectrum off the grating and focuses the beams onto vertically distinct positions of the 2D SLM.

The 2D SLM is a phase only device that can independently control the frequencydependent phase $\phi_i(\omega)$ of each beam *i*. By applying a phase in the frequency domain $E_i^0(\omega)$, the temporal profile of the ultrafast pulse $E'_i(t)$ can be modified. The relation between these quantities is given by the inverse Fourier transform:

$$E'_{i}(t) = \int_{-\infty}^{\infty} E^{0}_{i}(\omega) \exp\left[-i\phi_{i}(\omega)\right] \exp\left[i\omega t\right] d\omega$$
(4.4)

The global phase, time delay, and first-order dispersion (temporal chirp) can be controlled by applying a constant ($\phi_i = c_i$), linear ($\phi_i (\omega) = \omega t_i$), or quadratic phase ($\phi_i (\omega) = \omega^2 t_i^2$) in the frequency domain. Figure 4.5 illustrates the affect of applying different phase functions in the frequency domain on the temporal profile of the pulse.

Due to the pixelation of the SLM (480 × 480), the phase function applied to the spectral profile of the ultrafast pulse is not continuous. Similarly, the maximum grayscale value of the SLM (256) defines the number of phase values that can be applied at a given pixel. Because $\exp[i \times 0] = \exp[i \times 2\pi]$, we are able to wrap the applied phase modulo 2π . Thus, the 256 grayscale values need only apply a phase from 0 to 2π ; however in practice, phases of 0 to approximately 2.5π are applicable.

The discreteness of the applied phase function due to the pixelation of the SLM has a number of important implications. First, there is an inherent upper bound to the time delay that can be applied to the pulse. As discussed above, a time delay results from a linear change in the phase as a function of frequency. The time delay is thus defined as:

$$t_i = \frac{\partial \phi_i\left(\omega\right)}{\partial \omega} \tag{4.5}$$



Figure 4.5: Temporal pulse shaping. (a) Spectral profile of the fundamental pulse. (b) Temporal profile of the fundamental pulse. (c) Constant phase applied to the pulse. (d) Temporal profile of the constant phase pulse. (e) Linear phase applied to the pulse. (f) Temporal profile of the delayed pulse. (g) Quadratic phase applied to the pulse. (h) Temporal profile of the chirped pulse.

where $\partial \phi_i(\omega)$ is the change in phase and $\partial \omega$ is the frequency bandwidth per pixel, defined by the grating dispersiveness and the cylindrical lens focal length. To maximize the time delay, the change in phase from pixel to pixel must be maximized. The maximum applied phase is always 2π , which for a frequency bandwidth of approximately 0.03 THz per pixel, results in a maximum time delay of approximately 3 ps.

$$t_i^{max} = \frac{\partial \phi_i^{max}(\omega)}{\partial \omega} = \frac{2\pi}{2\pi \times 0.03 THz} \sim 3ps \tag{4.6}$$

To increase the upper limit of the time delay, the frequency bandwidth per pixel must be decreased. However, the timescales relevant to this thesis are of electronic coherences with dephasing times on the order of hundreds of femtoseconds. Thus, a maximum time delay of 3 ps is more than sufficient.

Second, the top-hat spatial profile of the pixels creates a periodic filter in the frequency domain that causes a rapid decline in the intensity of the shaped pulse as a function of the time delay. This pulse intensity roll off is given by the Fourier transform of the spatiallyfiltered frequency spectrum and results in a $sinc [\pi \partial \omega t]$ intensity decline, where $\partial \omega$ is the frequency bandwidth per pixel. Similarly, the finite spectral resolution of the grating-cylindrical lens pair, $\Delta \omega$, generates a Gaussian time window $\exp[-\pi^2 \Delta \omega^2 t^2]$ that also attenuates the intensity of the shaped pulse [129].

Third, the pixelation of the SLM generates a train of pulses that are separated in time by $1/\partial \omega$. These sampling replica pulses generally occur at long time delays and their amplitudes decrease as a function of time due to the Gaussian time window and sinc function just described. A train of modulator replica pulses (separated in time by the applied time delay t) is also generated due to phase wrapping at modulo 2π and the periodic pixel boundaries. In general, modulator replica pulses are problematic for all coherent, ultrafast spectroscopy experiments using temporal pulse shaping.

4.3.1.3 Diffraction-Based Pulse Shaping

To separate the desired pulse from the modulator replica pulses, we employ a diffractionbased scheme of pulse shaping in which a sawtooth diffraction grating is applied along the vertical dimension of the 2D SLM [130]. The spatial phase (i.e. vertical position) and the amplitude of the sawtooth grating can be used to modify the frequency- and spatiallydependent phase:

$$\Phi(\omega, y) = \phi^{max} \left[\frac{1}{2} + A(\omega) S_d(\phi(\omega), y) \right]$$
(4.7)

where $\phi^{max} = 2\pi$ is the maximum achievable phase shift of the SLM, $S_d(\phi(\omega), y)$ is a sawtooth grating function of period d along the y dimension, and $A(\omega)$ is a frequencydependent amplitude function that can vary between 0 and 1. In this manner, the frequencydependent phase and amplitude of the output electric field can be modified and takes the form:

$$E_{i}^{\prime}(\omega) \propto E_{i}^{0}(\omega) \exp\left[-i\phi_{i}(\omega)\right] sinc\left[\pi - \frac{\phi^{max}}{2}A_{i}(\omega)\right]$$
(4.8)

where $E_i^0(\omega)$ is the input electric field.

A lens at the back focal plane of the SLM focuses each diffraction order off the sawtooth grating to a vertically distinct position where the first-order alone can be picked off by a mirror and collimated. By definition, each modulator replica pulse in the train has an additional phase factor of 2π . Thus, each modulator replica pulse is diffracted into a different order and only the desired pulse diffracts into the first-order. Figure 4.6 illustrates the scheme of diffraction-based pulse shaping and the applied sawtooth grating pattern.

The properties of the sawtooth grating pattern also define the minimum time delay achievable by the SLM. The minimum time delay is achieved when the phase across one period of the vertical sawtooth grating is minimized. The minimum applied phase is the maximum phase divided by the periodicity of the vertical sawtooth grating d, usually 6 pixels. However, pixels can also be binned in order to apply smaller phase changes over the spectral profile of the pulse. If the pixels are binned into 2 sets of 240 pixels, the minimum



Figure 4.6: Diffraction-based pulse shaping. (a) Beams spectrally dispersed by the grating (Gr) are focused by a cylindrical lens (CL) onto the spatial light modulator (SLM). (b) The sawtooth grating pattern applied along the vertical dimension $A(\omega) S_d(\phi(\omega), y)$ with a periodicity d diffracts the incoming beams into many vertically distinct spatial orders, eliminating modulator replica pulses and allowing amplitude shaping.

time delay achievable is 0.02 ps.

$$t_i^{min} = \frac{\partial \phi_i^{min}(\omega)}{\partial \omega} = \frac{(2\pi/d)/pixel}{\partial \omega} = \frac{(2\pi/6)/240}{2\pi \times 0.03THz} \sim 0.02ps$$
(4.9)

To decrease the time step lower limit, the frequency bandwidth per pixel or the periodicity of the vertical sawtooth grating can be increased.

Not only does diffraction-based pulse shaping have the distinct advantages of providing spectral amplitude control and elimination of unwanted modulator replica pulses, it also allows spectroscopic experiments to be performed in the rotating frame and for easy phase cycling, both of which increase the quality of measured spectroscopic signals. Rotating frame detection and phase cycling will be discussed at the end of the chapter.

4.3.2 Experimental Calibrations

In order to perform an ultrafast experiment, the SLM must be carefully calibrated to apply the correct frequency-dependent phase for each desired output field. Thus, the phase applied by a given grayscale value of the SLM and the frequency component of a pulse on a given pixel must be calibrated. Once the 2D SLM has been calibrated, the compression and time ordering of the pulses must be set for time-resolved experiments. Finally, the pulse intensity roll off, carrier frequency, and global phase difference between the signal and local



Figure 4.7: Phase-to-grayscale calibration. (a) The input beam reflects off a beamsplitter (BS) and diffracts off a linear phase mask (PM) to produce two beams. The beams are focused onto horizontally distinct regions of the 2D spatial light modulator (SLM). The phase of one beam is modified by changing the grayscale value of the SLM $\phi(x)$, while the phase of the other beam is held constant $\phi(0)$. The beams reflect off the SLM and are refocused onto the PM by the CL and are recombined. The backward propagating beams are transmitted through the BS and the interference between the beams is measured on a photodiode (PD). (b) Interference of phase-perturbed beams. (c) Extracted phase-to-grayscale calibration function.

oscillator electric fields must be measured.

4.3.2.1 Phase-to-Grayscale

First, the phase applied by a given grayscale value of the SLM must be calibrated. To do this, one beam is reflected off a beam splitter and onto a 1D phase mask. Diffraction off the phase mask produces two first-order beams that are focused by a cylindrical lens onto two horizontally distinct regions of the 2D SLM. The phase of one beam is modified by changing the grayscale value of the SLM $\phi(x)$, while the phase of the other beam is held constant $\phi(0)$. The phase-perturbed and phase-constant beams reflect off the SLM and are refocused by the cylindrical lens onto the phase mask and are recombined. The backward propagating beams are transmitted through the beam splitter and their interference is measured on a photodiode as a function of the SLM grayscale value, $\cos[\phi(x)]$. From this measurement, a phaseto-grayscale calibration function can be determined to apply the correct phase to a given frequency component. Figure 4.7 illustrates the experimental setup, the raw data, and the extracted phase-to-grayscale calibration function. Once the phase-to-grayscale calibration has been completed, recalibration is not necessary unless there is a significant change in wavelength of the beam due to the frequency-dependence of the applied phase by a given



Figure 4.8: Pixel-to-wavelength calibration. (a) The three pixel-wide narrowband spectrum of the ultrafast pulse is measured across all pixels of the spatial light modulator. (b) Extracted pixel-to-wavelength calibration function.

grayscale value.

4.3.2.2 Pixel-to-Wavelength

In order to control the frequency-dependent phase of an ultrafast pulse, the pixel on which a given frequency component is located must be measured. To do this, a three pixel-wide column of the sawtooth grating pattern is illuminated on the SLM. Only the frequency components on the illuminated pixels diffract, exit the pulse shaper, and are measured by the spectrometer. By sweeping the location of the illuminated pixels across the SLM and measuring the corresponding spectra, a pixel-to-wavelength calibration function can be determine. Figure 4.8 illustrates the experimental concept and an extracted pixel-towavelength calibration function. As the frequency components of the pulse and the alignment of the beam change daily, the pixel-to-wavelength calibration must also be performed daily.

4.3.2.3 Pulse Compression and Timings

In ultrafast experiments, the compression and time ordering of pulses are critical. Propagation through transmissive optics and off the grating, cylindrical lens, and SLM (which acts as



Figure 4.9: Pulse compression and timings. (a) In the cross correlation setup, two beams are focused by lens L1 into a Beta Barium Borate (BBO) crystal for second harmonic generation and are recollimated by lens L2. A bandpass filter (BPF) blocks out the fundamental light and an iris (I) isolates the cross correlation signal, which is measured by a photodiode (PD) as a function of the time delay between the two beams. (b) Cross sectional image of the four fundamental beams and all the second harmonic cross correlations.

a grating compressor) changes the compression and timing of the pulses. So, the compression and timing must be calibrated daily. First, the pulse duration of one beam is measured by second harmonic frequency-resolved optical grating (SHG FROG) and compressed using the grating and cylindrical lens in the pulse shaper [131]. Next, a BBO crystal is placed at the sample position and SHG cross-correlations of each beam with the first are measured on a photodiode as a function of time delay between the beams. The SLM is used to compress the pulses and to overlap them in time by applying quadratic and linear phase, respectively. Figure 4.9 illustrates the cross correlation setup and the second harmonic cross correlations of the four fundamental beams.

4.3.2.4 Pulse Intensity Roll Off

As discussed above, the intensity of a temporally shaped pulse decreases as a function of the applied time delay. The pulse intensity roll off can be calibrated by measuring the pulse spectrum as a function of time delay, $I(\tau)$. Figure 4.10 shows a frequency-integrated pulse in-



Figure 4.10: Pulse intensity roll off calibration. Frequency-integrated pulse intensity as a function of time delay, $I(\tau)$.

tensity roll off measurement. The pulse intensity roll off calibration can be used to correct for the decrease in the pulse intensity during a time-resolved experiment by dividing the 2D signal $S(\tau, 0, \omega_t)$ by the square root of the intensity roll off: $S^{IC}(\tau, 0, \omega_t) = S(\tau, 0, \omega_t) / \sqrt{I(\tau)}$. Once the pulse intensity roll off has been measured, recalibration is not necessary unless there is a significant change in the frequency bandwidth per pixel.

4.3.2.5 Carrier Frequency

The precise carrier frequency of each beam must also be measured because our experiments are inherently performed in the rotating frame. To determine the carrier frequencies, each beam is temporally shaped to generate a symmetric double pulse by applying the phase function $\phi(\omega) = \cos[(\omega - \omega_c)T]$ and the interference of the two pulses is measured as a function of the delay time T by a spectrometer. The phase of the carrier frequency ω_c does not change as a function of time delay and thus its corresponding frequency fringe does not move in the interferogram. Once measured, the interferogram is inverse Fourier transformed into the time-time domain and the negative time component is filtered out. The interferogram is subsequently Fourier transformed back into the frequency-time domain to recover the electric field amplitude and phase and determine the carrier frequency. This data processing technique, called spectral interferometry, will be discussed in more detail



Figure 4.11: Carrier frequency calibration. (a) A symmetric double pulse is generated by applying the phase function $\phi(\omega) = \cos[(\omega - \omega_c)T]$ in the frequency domain. (b) The interference of the symmetric double pulse is measured as a function of T by a spectrometer. The fringe at the carrier frequency ω_c does not move as a function of the delay time T. (c) The interferogram is inverse Fourier transformed into the time-time domain. (d) The negative time component is filtered out. (e) and (f) The interferogram is Fourier transformed back into the frequency-time domain to recover the amplitude and phase of the electric field and determine the carrier frequency, which does not evolve as a function of the time delay.



Figure 4.12: Global phase calibration. (a) Interference between the signal and local oscillator fields is measured by a spectrometer as a function of the local oscillator phase $\phi_{LO}(x)$. (b) Local oscillator phase-dependent interference. (c) The global phase difference between the signal and local oscillator fields $\partial \phi$ is extracted from the frequency-integrated interference.

at the end of the chapter. Figure 4.11 illustrates the experimental concept, the raw data, and the extraction process. Like the pixel-to-wavelength calibration, daily changes in the alignment change the carrier frequency and thus the carrier frequency calibration must be performed daily.

4.3.2.6 Global Phase

To separate the absorptive and dispersive components of a spectroscopic signal, the phase difference between the signal and reference (i.e. local oscillator) beams must be determined. The global phase difference is calibrated by measuring the change in the signal-local oscillator interference as a function of the local oscillator phase $\phi_{LO}(x)$. Once the interferogram has been measured by a spectrometer, it is frequency integrated and fit to a $\cos [\phi_{LO}(x) + \partial \phi]$ function from which the global phase difference $\partial \phi$ can be determined. Figure 4.12 illustrates the experimental concept, the interferogram, and the extracted phase. Daily changes in the compression and time ordering of the pulses change the global phase difference between the signal and the local oscillator, thus the global phase calibration must be performed daily.

4.4 Spectroscopic Detection

4.4.1 Rotating Frame Detection

In temporal pulse shaping, a carrier frequency ω_c is used as a reference to control the phases of the other frequency components in the pulse. When the phase of a given frequency component is temporally shaped, it is modified relative to the carrier frequency. Thus, the phase of the carrier frequency remains constant as the phase of the other frequency components change. This subtle experimental detail has an important implication. Because the phase of the carrier frequency does not change, measurements are implicitly made in the rotating frame [124]. All oscillations measured experimentally are reduced by the carrier frequency, $\Omega = \omega - \omega_c$. Experimentally, we can thus define the Nyquist frequency for capturing the dynamics of a system and significantly reduce the required number of time steps per spectrum. This significantly reduces the time required to perform a full 2D measurement to approximately 15 minutes.

4.4.2 Spectral Interferometry

Spectral interferometry is a powerful tool used to characterize the electric field amplitude and phase of a spectrally resolved, non-self-heterodyned spectroscopic signal [125]. In this technique, a local oscillator $E_{LO}(\omega)$ at a time delay t_{LO} relative to the spectroscopic signal $E_{Signal}(\omega)$ is used as a reference for heterodyne detection [132]. The intensity of the total spectral interferometry signal $(I_{SI}(\omega, t_{LO}))$ is spectrally resolved by a spectrometer:

$$I_{SI}(\omega, t_{LO}) = |E_{LO}(\omega) \exp(i\omega t_{LO}) + E_{Signal}(\omega)|^{2}$$

$$= I_{LO}(\omega) + I_{Signal}(\omega) + E_{LO}(\omega) E_{Signal}^{*}(\omega) \exp(i\omega t_{LO})$$

$$+ E_{LO}^{*}(\omega) E_{Signal}(\omega) \exp(-i\omega t_{LO})$$
(4.10)



Figure 4.13: Spectral interferometry. (a) Spectral interferometry interference term in the frequency domain. (b) Interference term in the time domain. (c) Filtered interference term and heaviside step function in the time domain. (d) Signal amplitude in the frequency domain. (e) Signal phase in the frequency domain.

The interference term is isolated by subtracting the intensities of the local oscillator and signal from the total spectral interferometry signal:

$$\Delta I_{SI}(\omega, t_{LO}) = I_{SI}(\omega, t_{LO}) - I_{LO}(\omega) - I(\omega)$$

= $E_{LO}(\omega) E^*_{Signal}(\omega) \exp(i\omega t_{LO}) + E^*_{LO}(\omega) E_{Signal}(\omega) \exp(-i\omega t_{LO})$
= $2\Re [E^*_{LO}(\omega) E_{Signal}(\omega) \exp(-i\omega t_{LO})]$ (4.11)

Noting $C + C^* = (C_{\Re} + iC_{\Im}) + (C_{\Re} - iC) = 2C_{\Re}$.

When the interference term of the spectral interferometry signal is inverse Fourier transformed, the time domain signal is compromised of two signals at $\pm t_{LO}/2$.

$$IFT \left\{ \Re \left[E_{LO}^{*}(\omega) E_{Signal}(\omega) \exp(-i\omega t_{LO}) \right] \right\} = E_{LO}^{*}(t - t_{LO}/2) E_{Signal}(t - t_{LO}/2) + E_{LO}^{*}(-t - t_{LO}/2) E_{Signal}(-t - t_{LO}/2)$$
(4.12)

Filtering out the negative time component with a heaviside step function $\theta(t)$ to enforce causality (i.e. emission occurs after excitation), Fourier transformation of the filtered interference term back into the frequency domain results in a complex spectrum because the complex conjugate of the spectral interferometry signal was filtered out in the time domain. Multiplication of the filtered interference term in the time domain by the phase factor $\exp(i\omega t_{LO})$ corrects for the local oscillator time delay and division by the complex conjugate of the local oscillator electric field $E_{LO}^*(\omega)$ yields the amplitude and phase of the signal electric field:

$$E_{Signal}(\omega,\phi) = \frac{FT\left\{\theta\left(t\right) \times IFT\left\{\Re\left[E_{LO}^{*}\left(\omega\right)E_{Signal}\left(\omega\right)\exp\left(-i\omega t_{LO}\right)\right]\right\}\right\}\exp\left(i\omega t_{LO}\right)}{E_{LO}^{*}\left(\omega\right)}$$
(4.13)

This method for determining the amplitude and phase of a signal electric field is illustrated in figure 4.13 on the preceding page. In this case, a signal and local oscillator with Gaussian spectral profiles are used.

4.4.3 Phase Cycling

Phase cycling plays an extremely important role in 2D spectroscopy. Since experimental signals tend to be weak, phase cycling helps to suppress scatter from the exciting beams and enhance the desired signal contribution [126]. Phase cycling also performs the first step in spectral interferometry by subtracting out the local oscillator and signal intensity terms in order to retrieve the full amplitude and phase of the signal electric field.

In general, the total signal of a four beam experiment is the absolute value squared of the local oscillator, third-order signal, and three excitation electric fields:

$$I_{tot}(\omega) = |E_{LO}(\omega) \exp[i\phi_{LO}(\omega)] + E_{Sig}(\omega) \exp[i\phi_{Sig}(\omega)] + E_1(\omega) \exp[i\phi_1(\omega)] + E_2(\omega) \exp[i\phi_2(\omega)] + E_3(\omega) \exp[i\phi_3(\omega)]|^2 = I_{LO} + I_{Sig} + I_1 + I_2 + I_3 + E_{LO}E_{Sig} \cos[\phi_{LO} - \phi_{Sig}] + E_{LO}E_1 \cos[\phi_{LO} - \phi_1] + E_{LO}E_2 \cos[\phi_{LO} - \phi_2] + E_{LO}E_3 \cos[\phi_{LO} - \phi_3] + E_{Sig}E_1 \cos[\phi_{Sig} - \phi_1] + E_{Sig}E_2 \cos[\phi_{Sig} - \phi_2] + E_{Sig}E_3 \cos[\phi_{Sig} - \phi_3] + E_1E_2 \cos[\phi_1 - \phi_2] + E_1E_3 \cos[\phi_1 - \phi_3] + E_2E_3 \cos[\phi_2 - \phi_3]$$
(4.14)

where $\phi_{Sig}(\omega) = \pm \phi_1(\omega) \pm \phi_2(\omega) \pm \phi_3(\omega)$. To recover the electric field amplitude and phase of the third-order signal by spectral interferometry, the only desired signal contribution from equation 4.14 is the $E_{LO}E_{Sig}\cos[\phi_{LO} - \phi_{Sig}]$ term. To isolate this signal, phase cycling is used to subtract out the local oscillator and signal intensity terms (I_{LO} and I_{Sig}) and eliminate scattered contributions from the excitation beams.

First, we will consider a four-step phase cycling procedure in which the phase of each

excitation beam is individually shifted by π and subtracted from the unperturbed signal:

$$I^{4S}(\omega) = I_{tot} - I_{tot}(\phi_{LO}, \phi_1 + \pi, \phi_2, \phi_3)$$

$$-I_{tot}(\phi_{LO}, \phi_1, \phi_2 + \pi, \phi_3) - I_{tot}(\phi_{LO}, \phi_1, \phi_2, \phi_3 + \pi)$$

$$= -2I_{LO} - 2I_{Sig} - 2I_1 - 2I_2 - 2I_3 + 8E_{LO}E_{Sig}\cos[\phi_{LO} - \phi_{Sig}]$$

$$+4E_{Sig}E_1\cos[\phi_{Sig} - \phi_1] + 4E_{Sig}E_2\cos[\phi_{Sig} - \phi_2] + 4E_{Sig}E_3\cos[\phi_{Sig} - \phi_3]$$

$$+4E_1E_2\cos[\phi_1 - \phi_2] + 4E_1E_3\cos[\phi_1 - \phi_3] + 4E_2E_3\cos[\phi_2 - \phi_3]$$
(4.15)

With four-step phase cycling, the $E_{LO}E_i \cos [\phi_{LO} - \phi_i]$ terms, where i = 1, 2, 3, are eliminated and the desired signal contribution is enhanced by a factor of 8.

To eliminate additional undesired signal contributions, the logical extension of four-step phase cycling is a seven-step phase cycling procedure in which the phase of each pair of excitation beams is shifted by π and added to the four-step phase cycling signal.

$$I^{7S}(\omega) = I^{4S} + I_{tot}(\phi_{LO}, \phi_1 + \pi, \phi_2 + \pi, \phi_3) + I_{tot}(\phi_{LO}, \phi_1 + \pi, \phi_2 + \pi, \phi_3) + I_{tot}(\phi_{LO}, \phi_1, \phi_2 + \pi, \phi_3 + \pi) = I_{LO} + I_{Sig} + I_1 + I_2 + I_3 + 14E_{LO}E_{Sig}\cos[\phi_{LO} - \phi_{Sig}] -2E_{LO}E_1\cos[\phi_{LO} - \phi_1] - 2E_{LO}E_2\cos[\phi_{LO} - \phi_2] - 2E_{LO}E_3\cos[\phi_{LO} - \phi_3] +2E_{Sig}E_1\cos[\phi_{Sig} - \phi_1] + 2E_{Sig}E_2\cos[\phi_{Sig} - \phi_2] + 2E_{Sig}E_3\cos[\phi_{Sig} - \phi_3] +2E_1E_2\cos[\phi_1 - \phi_2] + 2E_1E_3\cos[\phi_1 - \phi_3] + 2E_2E_3\cos[\phi_2 - \phi_3]$$
(4.16)

Equation 4.16 reveals that seven-step phase cycling increases the number of undesired contributions but also increases the enhancement of the desired signal term to a factor of 14.

Adding an eighth phase cycling step in which the phases of all the excitation beams are shifted by π and subtracted from the seven-step phase cycling signal, the desired signal contribution can be isolated.

$$I^{8S}(\omega) = I^{7S} - I_{tot}(\phi_{LO}, \phi_1 + \pi, \phi_2 + \pi, \phi_3 + \pi)$$

= $16E_{LO}E_{Sig}\cos[\phi_{LO} - \phi_{Sig}]$ (4.17)

Indeed, the eight-step phase cycling procedure subtracts out the local oscillator and signal intensity terms, eliminates scatter from the excitation beams, and enhances the desired signal contribution by a factor of 16.

4.4.4 Phase Stability

Using common path optics and 2D temporal pulse shaping, the phases of an arbitrary number and geometry of beams are passively stabilized [133]. In our experimental setup, the phase stability is better than $\lambda/100$ over more than several hours. Strong phase stability between all beams is a distinct advantage of our 2D spectrometer. The phase stability allows us to perform two-quantum, three-dimensional, and higher-order measurements that most conventional 2D spectrometers are incapable of performing.

Chapter 5

Morphology and Excitonic Coupling in Double-Walled J-Aggregate Nanotubes

In chapter 5, we characterize the coupling between the inner and outer wall excitons of double-walled J-aggregate nanotubes. For our experiments, we take great care to ensure our samples are comprised of only isolated or bundled nanotubes in order to determine the effects of higher-order morphology on excitonic coupling. First, linear absorption spectra are correlated to morphology using cryogenic electron tunneling microscopy. Then, we use linear absorption to characterize nanotube morphology and measure two-dimensional spectra of isolated, bundled, and mixed nanotubes in solution. Finally, we characterize mixed nanotube samples in glass at cryogenic temperature.

Synthesis of the solution-based nanotubes was performed at MIT by Dörthe Eisele in the group of Moungi Bawendi. Synthesis of the glass nanotubes was also performed at MIT by Justin Caram and Sandra Doria in the Bawendi group. Cryogenic tunneling electron microscopy was performed at Brandeis University by Xiaofeng Fu in the group of Daniela Nicastro. Two-dimensional spectroscopy of solution-based nanotubes was performed by myself and Dylan Arias; two-dimensional spectroscopy of glass nanotubes was performed by myself and Jake Siegel.

5.1 Theory of Molecular Aggregates

The simplest exciton in a molecular aggregate is formed by a dimer coupled through its transition dipole moments [134, 38]. The Hamiltonian for the dimeric exciton states is determined by the lowest lying excited state energy of the molecular constituents E and the dipole-dipole coupling strength J:

$$H = \begin{bmatrix} E & J \\ J & E \end{bmatrix}$$
(5.1)

$$J = \frac{\mu_1 \mu_2}{4\pi \epsilon_r R^3} \left(1 - 3\cos^2 \theta \right)$$
(5.2)

The coupling strength is defined by the relative permittivity of the molecular constituents ϵ_r , the magnitudes of their transition dipole moments μ_i , and the distance R and angle θ between them, as illustrated in figure 5.1a. Coupling of the excited state molecular orbitals produces two Frenkel exciton states where the lower energy (i.e. bonding-like) wavefunction is the sum of the molecular orbitals and the higher energy (i.e. antibonding-like) wavefunction is the difference. The exciton state with in-phase molecular wavefunctions contains most of the oscillator strength because the individual dipoles act as an enhanced single dipole moment, while the exciton state with out-of-phase molecular wavefunctions contains little oscillator strength due to destructive interference of the dipole moments.

Coupling of the excited molecular states is strongly dependent on the relative angle between the transition dipole moments, shown in green in figure 5.1b. When $\theta = 0$, the coupling between the molecules is negative and the lowest lying energy state is characterized by inphase wavefunctions and dipoles aligned head-to-tail. The exciton absorption is red-shifted from the molecular excited state energy and the dimer is called a J-aggregate, illustrated in figure 5.1c. As the angle between the dipoles increases and dipole-dipole coupling becomes positive, the wavefunctions of the higher lying energy state become in-phase and the dipoles align side-by-side. When $\theta = \pi/2$, the exciton absorption is blue-shifted from the molecular



Figure 5.1: Coupled molecular dimer. (a) Transition dipole moments μ_1 and μ_2 of molecules 1 and 2 are separated by R and oriented at angle θ . (b) Angle-dependent dipole-dipole coupling and eigenstate energies. (c) Molecular dimer wavefunctions for the J- and H-aggregates ψ_{\pm}^J and ψ_{\pm}^H , respectively.



Figure 5.2: Linear J-aggregate. Linear J-aggregates possess molecular dipole moments oriented parallel to one another in a chain, which are separated by \overrightarrow{r} and have a nearestneighbor interaction strength J.

excited state energy and the dimer is called an H-aggregate, illustrated in figure 5.1c.

5.1.1 Linear J-Aggregates

The coupled dimer model can be easily extended to linear J-aggregates where many twolevel molecules are aligned in one dimension and couple to form delocalized exciton states, illustrated in figure 5.2 [135]. The Frenkel exciton Hamiltonian takes the form:

$$\hat{H} = \sum_{p} E_p B_p^{\dagger} B_p + \sum_{p,q} V_{pq} B_p^{\dagger} B_q$$
(5.3)

$$V_{pq} = -\frac{J}{\left|\overrightarrow{r}_{p} - \overrightarrow{r}_{q}\right|^{3}}$$

$$(5.4)$$

where E_p is the excited state molecular orbital energy of molecule p, B_p^{\dagger} and B_p are the creation and annihilation operators for an exciton on molecule p (under the constraint there can be at most one exciton per molecule), and V_{pq} is the dipole-dipole coupling between molecules p and q at \overrightarrow{r}_p and \overrightarrow{r}_q with nearest-neighbor interaction strength J.

In the idealized linear model, J-aggregates are homogeneous and intermolecular coupling is restricted to nearest neighbors. Under this condition, the solutions to the exciton wavefunctions, energies, and transition dipole intensities take the form:

$$|\Psi_k\rangle = \sqrt{\frac{2}{N+1}} \sum_p \sin\left[\frac{kp\pi}{N+1}\right] |\psi_p\rangle \tag{5.5}$$

$$E_k = E + 2J \cos\left[\frac{k\pi}{N+1}\right] \tag{5.6}$$

$$I_{k} = \left(1 - (-1)^{k}\right) \frac{\mu^{2}}{N+1} \cot^{2} \left[\frac{k\pi}{2(N+1)}\right]$$
(5.7)

where k is the crystal wavevector, N is the number of coupled molecules in the J-aggregate, and $|\psi_p\rangle$ are the molecular wavefunctions of molecule p. Similar to particle-in-a-box states, the J-aggregate wavefunctions have a sinusoidal form that is characterized by the exciton delocalization. Large spatial delocalization is favorable and substantially stabilizes the exciton by lowering the J-aggregate energy. It also substantially increases the macroscopic transition dipole moment.

5.1.2 J-Aggregate Nanotubes

In some cases, the physical structure of molecules allows complex aggregate structures to form in solution, including sheets, rods, and ribbons. In chapters 5 and 6, we study J-aggregates that form double-walled nanotubes [136, 137, 138]. Such nanotubes can be modeled by wrapping a conventional planar J-aggregate into a cylinder of the appropriate radius, illustrated in figure 5.3a. In this model, molecules of length a and thickness d are arranged in a brick lattice defined by the basis vectors \vec{a}_1 and \vec{a}_2 , where the displacing shift s between molecules in adjacent rows determines the dipole-dipole coupling and the type of aggregation. The planar structure is then wrapped along a wrapping vector $\vec{C} = n_1 \vec{a}_1 + n_2 \vec{a}_2$, where n_i is an integer, to create a cylinder of the appropriate radius R, where $\left|\vec{C}\right| = 2\pi R$ by normal Euclidean geometry. The vector \vec{Z} defines the long axis of the cylinder and θ the angle between the molecular transition dipole moments and the long axis [139, 140].

In the theoretical description of J-aggregate nanotubes, each molecule is characterized by a vector defining its transition dipole moment $\overrightarrow{\mu}$. The Frenkel Hamiltonian of the J-



Figure 5.3: J-aggregate nanotubes. (a) Brick lattice defined by the basis vectors \overrightarrow{a}_1 and \overrightarrow{a}_2 , where *a* is the length, *d* the thickness, and *s* the displacing shift between molecules. The planar aggregate is wrapped along the wrapping vector $\overrightarrow{C} = n_1 \overrightarrow{a}_1 + n_2 \overrightarrow{a}_2$, where n_i is an integer, to create a cylinder of the appropriate radius *R* and \overrightarrow{Z} defines the long axis of the cylinder and θ the angle between the molecular transition dipole moments $\overrightarrow{\mu}$ and the long axis. (b) J-aggregate nanotube composed of a stack of N_1 equidistant rings containing N_2 equidistant molecules and characterized by a helical angle γ between rings.

aggregate nanotube has the familiar form:

$$\hat{H} = \sum_{p} E_p B_p^{\dagger} B_p + \sum_{p,q} V_{pq} B_p^{\dagger} B_q$$
(5.8)

$$V_{pq} = \frac{\overrightarrow{\mu}_{p} \cdot \overrightarrow{\mu}_{q}}{\left|\overrightarrow{r}_{p} - \overrightarrow{r}_{q}\right|^{3}} - 3 \frac{\left(\overrightarrow{\mu}_{p} \cdot \overrightarrow{r}_{pq}\right)\left(\overrightarrow{\mu}_{q} \cdot \overrightarrow{r}_{pq}\right)}{\left|\overrightarrow{r}_{p} - \overrightarrow{r}_{q}\right|^{5}}$$
(5.9)

where V_{pq} is the dipole-dipole coupling and is dependent on the distance $\overrightarrow{r}_p - \overrightarrow{r}_q$ and orientation between the dipole moments $\overrightarrow{\mu}_p$ and $\overrightarrow{\mu}_q$ of molecules p and q, respectively. The exciton wavefunction take the form:

$$|\Psi_k\rangle = \sum_p \psi_k(p) B_p^{\dagger} |g\rangle$$
(5.10)

where k is a general quantum number, $\psi_k(p)$ are the single exciton eigenvectors, and $|g\rangle$ is the exciton ground state.

J-aggregate nanotubes can be modeled equivalently by a stack of N_1 equidistant rings containing N_2 equidistant molecules with each ring rotated by a helical angle γ , illustrated in figure 5.3b. In this case, the Hamiltonian is separable into a one-dimensional longitudinal contribution of the N_1 rings (with the same solutions of the linear J-aggregate model described in equations 5.5-5.7) and radial contributions from the independent rings with Bloch wavefunctions:

$$|\Psi_{k_2}\rangle = \exp\left[i2\pi k_2 n_2/N_2\right]$$
 (5.11)

where $k_2 = 0, \pm 1, \ldots, N_2/2$ and n_2 is the position of a molecule in the ring [141]. The three-dimensional structure of the nanotube produces three excitons bands $K = (k_1, k_2)$: a longitudinal band characterized by wavevector k_1 and two degenerate radial bands characterized by wavevectors $\pm k_2$. Moreover, their linear absorption spectra are dominated by these three K states: (1,0) a superradiant state polarized along the long axis and $(0, \pm 1)$ two degenerate states polarized perpendicular to the long axis.

5.2 Double-Walled J-Aggregate Nanotubes

In our experiments, we study J-aggregates formed by the self-assembly of 3,3'-bis(2-sulfopropyl) 5,5',6,6'-tetrachloro 1,1'-dioctylbenzimidacarbocyanine 18 (C₈S₃) monomers in water, illustrated in figure 5.4a. C₃S₈ monomers consist of three important structural areas [142]. The conjugated backbone of the isocyanine dye (dark grey) largely determines the electronic structure of the monomer, while the hydrophobic alkyl sidechains (light grey) and hydrophilic sulfate groups (red) create regions of differing hydrophobicity and drive self-assembly in water. C₃S₈ is first dissolved in methanol to prepare a monomer solution that exhibits broad absorption around 520 nm, shown in figure 5.4b.

5.2.1 Isolated Nanotubes

Upon dilution of the monomer solution to a final 9:1 water:methanol ratio, C_3S_8 selfassembles to form J-aggregate double-walled nanotubes, illustrated in figure 5.4c. The linear absorption of the J-aggregate nanotubes (figure 5.4b) exhibits the characteristic red-shift and



Figure 5.4: Isolated J-aggregate nanotubes. (a) 3,3'-bis(2-sulfopropyl) 5,5',6,6'-tetrachloro 1,1'-dioctylbenzimidacarbocyanine 18 (C₃S₈) monomers, composed of a conjugated cyanine backbone (dark grey), hydrophobic alkyl chains (light grey), hydrophilic sulfate groups (red), and sodium counter ions. (b) Monomer (black) and isolated nanotube (red) linear absorption. The 599 nm and 589 nm J-aggregate peaks correspond to excitons on the inner and outer walls, respectively. (c) Double-walled J-aggregate nanotube with characteristic inner and outer walls. (d) Cryogenic tunneling electron microscopy of isolated nanotubes reveals an inner wall diameter of 6 nm and an outer wall diameter of 13 nm (with a wall-to-wall separation of 3-4 nm) and lengths up to a few microns.

lineshape narrowing due to coupling of the electronic states when the isocyanine backbones stack in parallel [36, 143, 144, 145]. Upon aggregation, two distinct excitonic bands form at 599 nm and 589 nm, which have been previously determined to arise from inner and outer wall excitons, respectively [146]. Weaker, higher energy exciton bands also exist as discussed in the theoretical description of J-aggregate nanotubes. The most important of these higher energy exciton bands appears at 583 nm as a shoulder to the 589 nm peak. From cryogenic tunneling electron microscopy (cryo-TEM), the structure of isolated C_3S_8 nanotubes has been determined to consist of a 6 nm diameter inner wall and a 13 nm diameter outer wall separated by approximately 3-4 nm (corresponding to the alkyl chain length), shown in figure 5.4d. The nanotubes grow up to microns in length and are homogeneous along the entire length [147].

5.2.2 Bundled Nanotubes

Upon aging of isolated nanotubes, bundles of nanotubes form and exhibit their own characteristic linear absorption, shown in figure 5.5a. The inner wall peak lightly red-shifts to 603 nm and is slightly broadened. Additionally, a new broad, higher energy peak appears at 578 nm. Cryo-TEM, shown in figure 5.5b, reveals that bundled nanotube ensembles are far less homogeneous than isolated nanotubes. Bundles were composed of 6 to 12 nanotubes and appeared in two morphologies: straight and twisted, shown in figure 5.5d. Straight bundles were characterized by multiple nanotubes running parallel to one another along their entire length, while twisted bundles exhibited curvature along their length for which a unique twisting angle could be defined.

To determine the high-order morphology of the bundles, a series of TEM images was taken at different angles between the electron beam and sample in order to reconstruct the full three-dimensional structure. From the spatial reconstruction, a cross-sectional slice of a bundle can be well determined, one example of which is shown in figure 5.5d. The average center-to-center distance between nanotubes in a bundle is 9 nm, which is inconsistent with a



Figure 5.5: Bundled J-aggregate nanotubes. (a) Linear absorption of isolated (red) and bundled (blue) nanotubes. The 603 nm and 578 nm bundled nanotube peaks originate from inner wall excitons polarized parallel and perpendicular to the long axis of the bundle, respectively. (b) Cryogenic tunneling electron microscopy of bundled nanotubes. (c) Graphical representation of bundled inner wall nanotubes. (d) Three dimensional reconstructions of straight and twisted bundles, cross-sectional slice of a twisted bundle, and reconstruction of the cross-sectional slice.

bundle of double-walled nanotubes. However, the center-to-center distance is consistent with a bundle of inner wall nanotubes. In order to maintain its morphology, the bundled inner wall nanotubes would require an amorphous "outer wall" surrounding the entire bundle due to the hydrophobicity of the alkyl sidechains. Assuming there is substantial disorder in the amorphous outer wall, its spectral features should not exhibit the characteristic red-shifting and line narrowing of J-aggregation. Thus, we hypothesize the well-defined spectral features in the linear absorption spectrum of bundled nanotubes, shown in figure 5.5a, arise solely from inner wall nanotubes.

5.2.2.1 Linear Dichroism

To determine the transition dipole moment orientations in isolated and bundled nanotubes, we measured reduced linear dichroism spectra: $LD_R = (A_{\parallel} - A_{\perp}) / (A_{\parallel} + A_{\perp})$ [148]. In this experiment, nanotubes were flowed through a cell which caused their long axis to align along the direction of flow due to their large length to diameter ratio, illustrated in figure 5.6a. Absorption of light polarized parallel (A_{\parallel}) and perpendicular (A_{\perp}) to the flow direction (also labelled $H(\parallel)$ and $V(\perp)$, respectively) was then measured. A linear dichroism of 1 indicates a transition dipole moment polarized perfectly parallel to the flow direction and -1 indicates a transition perfectly perpendicular to the flow direction. A linear dichroism of 0 indicates a transition dipole moment that is either isotropic or polarized 45° relative to the flow direction.

The reduced linear dichroism of the isolated nanotubes shown in figure 5.6b demonstrates the strong inner and outer wall transitions at 599 nm and 589 nm, respectively, are polarized mainly parallel to the long axis of the nanotubes, while the small shoulder at 583 nm is polarized mainly perpendicular. The reduced linear dichroism of the bundled nanotubes shown in figure 5.6c demonstrates that the strong transition at 603 nm is polarized mainly parallel to the long axis of the bundles, while the weak transition at 578 nm is polarized mainly perpendicular .



Figure 5.6: Reduced linear dichroism. (a) Nanotubes flown through a cell align along the flow direction, parallel to the laser polarization H(||) and perpendicular to the laser polarization $V(\perp)$. (b) Reduced linear dichroism of isolated nanotubes. The two strong inner and outer wall transitions at 599 nm and 589 nm, respectively, are polarized mainly parallel to the long nanotube axis. A third, weak shoulder at 583 nm is polarized mainly perpendicular to the long axis of the nanotube. (c) Reduced linear dichroism of bundled nanotubes. The lower energy transition at 603 nm is polarized mainly perpendicular to the long axis of the bundles.


Figure 5.7: Polarization-controlled two-dimensional experiments. Co-linear experiments utilize pulses with parallel polarizations (H) to excite and detect excitons with parallel transition dipole moments, which enhances diagonal peaks. Cross-linear experiments utilize pulses with perpendicular polarizations to excite (V) and detect (H) excitons with perpendicular transition dipole moments, which enhances crosspeaks between perpendicularly polarized states. Isolated and bundled nanotube co-linear and cross-linear 2D spectra with the enhanced peaks indicated in grey.

5.3 Morphology and Excitonic Coupling

From linear absorption and linear dichroism, the structural origin and orientation of the spectral features in double-walled J-aggregate nanotubes has been determined. In the isolated nanotubes, the two strongest spectral features at 599 nm and 589 nm have been shown to originate from the inner and outer walls, respectively. Previous chemical oxidation experiments also demonstrated at most weak coupling between excitons on the inner and outer walls [146]. However, these experiments were only indirect measurements of exciton coupling. Thus, we use two-dimensional (2D) spectroscopy to directly measure the coupling of excitons in double-walled J-aggregate nanotubes.

In our experiments, we utilize polarization control to interrogate exciton states with different transition dipole moment orientations as illustrated in figure 5.7. Polarizationcontrolled experiments require orientated transition dipole moments, thus the nanotubes are flowed through a cell to align their long axes parallel to the direction of flow in the same manner as the linear dichroism experiments. We perform two types of polarizationcontrolled 2D experiments: co-linear and cross-linear. Co-linear 2D experiments use laser fields with all parallel polarizations to excite and detect excitons with the same transition dipole moment orientation, enhancing the diagonal peaks of excitons with transition dipole moments parallel to the laser field polarization. Cross-linear 2D experiments use pairs of laser fields with perpendicular polarizations to excite and detect excitons with perpendicular transition dipole moments, enhancing the crosspeaks between excitons with perpendicular transition dipole moments.

Co-linear and cross-linear 2D spectra in isolated and bundled nanotubes are illustrated in figure 5.7. The 2D measurements of isolated nanotubes interrogate two excitons with transition dipole moments parallel and one exciton perpendicular to the long axis of the nanotubes. The co-linear experiment interrogates the diagonal peaks of excitons with the parallel transition dipole moments and their crosspeaks. The cross-linear experiment interrogates the crosspeaks between excitons with perpendicular transition dipole moments. The 2D measurements of bundled nanotubes interrogate one exciton with a transition dipole moment parallel and one exciton perpendicular to the long axis of the bundles. The co-linear experiment interrogates the diagonal peak of the exciton with the parallel transition dipole moment. The cross-linear experiments interrogates the crosspeak between the excitons with perpendicular transition dipole moments. The co-linear and cross-linear spectra for isolated and bundled nanotubes are illustrated in figure 5.7 with a light grey box indicating the peak enhancement in each experimental case.

Many two-dimensional experiments have been performed on J-aggregate nanotubes previously [149, 47, 48, 150, 151, 49]. However, the appropriate care was not taken to ensure only a single nanotube morphology was interrogated in each of these studies. Because linear dichroism and linear absorption have proven to be reliable methods for determining nanotube morphology, we used both techniques throughout the synthesis and 2D experiments to ensure samples contained only the desired morphology.

5.3.1 Isolated Nanotubes

First, the co-linear 2D spectrum of isolated nanotubes was measured with the polarization of the incident laser fields parallel to the flow direction. The co-linear 2D spectrum in figure 5.8a reveals two diagonal peaks at 599 nm and 589 nm associated with the inner and outer wall excitons polarized parallel to the long axis of the nanotube. A weak crosspeak between the two peaks is also clearly visible, revealing weak inter-wall coupling between excitons on the inner and outer walls.

Second, the cross-linear 2D spectrum of isolated nanotubes was measured with the polarization of the exciting laser fields parallel and the detecting laser fields perpendicular to the flow direction. The cross-linear 2D spectrum in figure 5.8c reveals the same inner and outer wall diagonal peaks at 599 nm and 589 nm, respectively. However, now there is a much stronger crosspeak shifted to lower excitation wavelength. This strong crosspeak appears between the 599 nm inner wall peak and the weak 583 nm shoulder of the outer wall peak.

The reduced linear dichroism revealed this shoulder is polarized perpendicular to the long axis of the nanotube, explaining the appearance of the new crosspeak between the perpendicularly polarized exciton states due to the orthogonality of the incident laser fields. However, the 583 nm diagonal peak still does not appear because the exciting laser fields are perpendicular to the transition dipole moment. Additionally, the weak crosspeak between inner and outer wall excitons can still be observed as a small shoulder to higher excitation wavelength.

As discussed previously, the theoretical description of J-aggregate nanotubes determined that three exciton bands form due to their three-dimensional cylindrical structure: one perpendicular to the long axis and two perpendicular. In double-walled nanotubes, two sets of three exciton bands should exist and correspond to each wall. The 583 nm shoulder is



Figure 5.8: Isolated nanotube two-dimensional correlation spectra. (a) Co-linear 2D correlation spectrum of isolated nanotubes (enhancing the diagonal peaks) reveals weak coupling between inner and outer wall peaks at 599 nm and 589 nm. (b) Cross-linear 2D correlation spectrum of isolated nanotubes (enhancing the crosspeaks) reveals strong coupling between perpendicular inner wall peaks at 599 nm and 583 nm. (c) Co-linear 2D rephasing spectrum Feynman diagrams (stimulated emission only), where green and blue arrows denotes transitions at 599 nm and 589 nm, respectively, and solid and dashed arrows denote transition dipole moment interactions parallel and perpendicular to the long axis of the nanotubes, respectively. (d) Cross-linear 2D rephasing spectrum Feynman diagrams.

the lowest energy peak polarized perpendicular to the long axis of the nanotube and thus we expect it to originate from the same state as the lowest energy peak polarized parallel to the long axis, i.e. the inner wall. Strong exciton coupling between the 583 nm and 599 nm peaks also supports the assignment of the 583 nm peak to an inner wall exciton. The co-linear 2D spectrum revealed weak coupling between parallel inner and outer wall excitons, thus strong coupling is similarly unlikely between perpendicular excitons on different walls. So, we assign the 583 nm peak to an inner wall exciton polarized perpendicular to the long axis of the nanotube, revealing strong intra-wall coupling of excitons in isolated nanotubes.

5.3.2 Bundled Nanotubes

Third, the co-linear 2D spectrum of bundled nanotubes was measured with the polarization of the incident laser fields parallel to the flow direction. The co-linear spectrum in figure 5.9a reveals a strong diagonal peak at 603 nm and a weak diagonal peak at 578 nm, corresponding with the linear absorption of the bundled nanotubes. The linear dichroism revealed these peaks are polarized parallel and perpendicular to the long axis of the bundle, respectively, and are likely associated with excitons on the bundled inner walls. The transition dipole moment orientation and laser field polarization explains the relative amplitude of the two peaks in the co-linear 2D spectrum. A weak crosspeak between the two peaks is also observed.

Fourth, the cross-linear 2D spectrum of bundled nanotubes was measured with the polarization of the exciting laser fields parallel and the detecting laser fields perpendicular to the flow direction. The cross-linear 2D spectrum in figure 5.9c reveals the same 603 nm diagonal peak but with much weaker amplitude as the exciting laser fields are now perpendicular to the transition dipole moment. However, the crosspeak has much stronger amplitude as the detecting laser fields are now parallel to transition dipole moment of the 578 nm peak.

We assume both the 603 nm and 578 nm peaks originate from the inner wall and there is no spectral contribution from the amorphous outer wall. Thus in direct comparison to the cross-linear 2D spectrum of isolated nanotubes, the strong crosspeak in the bundled



Figure 5.9: Bundled nanotube two-dimensional correlation spectra. (a) Co-linear 2D correlation spectrum of bundled nanotubes (enhancing the diagonal peaks) reveals coupling between perpendicular inner wall excitons at 603 nm and 578 nm. (b) Cross-linear 2D correlation spectrum of bundled nanotubes (enhancing the crosspeaks) reveals strong coupling between perpendicular inner wall excitons at 603 nm and 578 nm. (c) Co-linear 2D rephasing spectrum Feynman diagrams (stimulated emission only), where solid and dashed arrows denote transition dipole moment interactions parallel and perpendicular to the long axis of the bundles, respectively. (d) Cross-linear 2D rephasing spectrum Feynman diagrams.



Figure 5.10: Mixed nanotube two-dimensional correlation spectrum. (a) Co-linear 2D correlation spectrum of mixed nanotubes (enhancing the diagonal peaks) with incident laser polarizations $P = 45^{\circ}$ between H(||) and $V(\perp)$. Co-linear 2D spectrum reveals no crosspeak between excitons on isolated nanotubes and bundled nanotubes, demonstrating they are distinct physical and electronic systems in solution. (b) Emission energy-integrated projection of co-linear 2D correlation spectrum onto the excitation axis.

nanotube spectrum also reveals strong coupling between intra-wall excitons.

5.3.3 Mixed Nanotubes

Finally, the co-linear 2D spectrum of a mixed sample of isolated and bundled nanotubes was measured with the polarization of the incident laser fields 45° relative to the flow direction, allowing contributions from all the spectral features. The co-linear 2D spectrum in figure 5.10 reveals three diagonal peaks. The 600 nm peak arises from excitons on the inner walls of isolated (599 nm) and bundled (603 nm) nanotubes polarized parallel to the long axes. The 589 nm peak arises primarily from excitons on the outer wall of isolated nanotubes. The 578 nm peak arises primarily from excitons on the inner walls of bundled nanotubes polarized perpendicular to the long axis of the bundle. Additionally, a single crosspeak is observed between the bundled nanotube diagonal peaks (578 nm, 603 nm). However, no crosspeaks are observed between excitons on isolated and bundled nanotubes, which clearly demonstrates the two morphologies are physically and electronically distinct systems in solution.



Figure 5.11: Glass J-aggregate nanotubes. (a) Glass (green) and solution-based isolated (red) nanotube linear absorption. Nanotubes are embedded in glass formed by a sucrose:trehalose matrix. Glass nanotubes exhibit lineshape broadening and contributions from isolated and bundled nanotubes. (b) Glass nanotubes are isotropically oriented in the micro-cuvette.

5.4 Double-Walled J-Aggregate Nanotubes at Low Temperature

To perform experiments at low temperature, J-aggregate nanotubes must be preserved in the solid state while maintaining their self-assembled structure, higher-order morphology, and optical and electronic properties. A supersaturated solution of a 1:1 mixture of sucrose and trehalose was used to prepare J-aggregate nanotubes in a glass, which is formed by a sugar matrix [152, 153, 154]. First, the solution-based nanotubes were diluted by 50% with the supersaturated sugar solution. Then, a thin layer of the nanotube-sugar solution was drop cast onto a micro-cuvette. Finally, the thin layer of the nanotube-sugar solution was exposed to vacuum for 24 hours to evaporate the water and form a glass with the nanotubes imbedded in the sugar matrix.

Sucrose and trehalose sugar matrices have been frequently used to form glasses of proteins and other biologically-relevant molecules while retaining their structure. The linear absorption of a J-aggregate nanotube glass shown in figure 5.11a demonstrates that the nanotubes maintain their optical properties upon glass formation. Compared to the linear absorption of isolated nanotubes in solution, the glass linewidths are slightly broadened and the higher wavelength shoulder of the inner wall peak at 599 nm indicates that some bundled nanotubes are also present. However, the self-assembled structure and higher-order morphology of the nanotubes is unlikely to change significantly in the glass since the linear absorption of the isolated nanotubes (figure 5.4c) changes drastically due to the small perturbation of bundling (figure 5.5a).

Additionally, J-aggregate nanotubes are not aligned during glass formation and are isotropically oriented in the micro-cuvette, illustrated in figure 5.11b. Thus, polarizationcontrolled experiments are not possible since alignment of the nanotubes and their transition dipole moments is required. Our collaborators in the Bawendi group are currently working on a method to orient the glass nanotubes by flowing the supersaturated sugar and nanotube solution during the evaporation step of the synthesis.

5.4.1 Exciton Dephasing

With J-aggregate nanotubes in the solid state, temperature-dependent experiments can easily be performed outside the range of 273 to 373 K. We used 2D rephasing spectroscopy to measure the temperature-dependent homogeneous (i.e. anti-diagonal) and inhomogeneous (i.e. diagonal) linewidths and investigate the mechanisms of exciton dephasing. First, we performed 2D experiments on glass nanotubes at 10 K and measured the linewidths of the inner and outer wall peaks of isolated nanotubes at 594 nm and 584 nm, respectively, and the inner wall peak of bundled nanotubes at 596 nm. The rephasing spectrum of the glass nanotubes is shown in figure 5.12a and the homogeneous and inhomogeneous linewidths are summarized in table 5.1. The rephasing spectrum at 10 K revealed inhomogeneously broadened lineshapes for all three peaks. At low temperature, few vibrational or phonon modes are populated due to low thermal energy, thus exciton dephasing is dominated by static disorder in the glass nanotubes.

Second, we performed 2D experiments on solution-based isolated nanotubes at 295 K and measured the linewidths of the inner and outer walls peaks at 597 nm and 587 nm,



Figure 5.12: Homogeneous and inhomogeneous linewidths at room and cryogenic temperatures. (a) Rephasing spectrum of glass nanotubes (isolated and bundled) at 10 K. (b) Diagonal line cut of the rephasing spectrum at 10 K. (c) Rephasing spectrum of solutionbased isolated nanotubes at 295 K. (d) Diagonal cut of the rephasing spectrum at 295 K.

Peak	Solution L	inewidths at 295 K	Doolz	Glass Linewidths at 10 K		
	Inhomog.	Homog.	1 eak	Inhomog.	Homog.	
587 nm	5.4 nm	4.6 nm	584 nm	3.0 nm	1.8 nm	
597 nm	4.2 nm	4.2 nm	594 nm	2.6 nm	1.5 nm	
	4.2 1111	4.2 1111	596 nm	2.8 nm	1.5 nm	

Linewidths at Cryogenic and Room Temperatures

Table 5.1: Homogeneous and inhomogeneous linewidths of J-aggregate nanotubes at room and cryogenic temperatures. Homogeneous and inhomogeneous linewidths of glass nanotubes at 10 K, including the inner and outer wall peaks of isolated nanotubes at 594 nm and 584 nm, respectively, and the inner wall peak of bundled nanotubes at 596 nm. Homogeneous and inhomogeneous linewidths of solution-based isolated nanotubes at 295 K, including the inner and outer wall peaks at 597 nm and 587 nm, respectively.

respectively. The rephasing spectrum of the isolated nanotubes is shown in figure 5.12c and the homogeneous and inhomogeneous linewidths are summarized in table 5.1. The rephasing spectrum at 295 K revealed homogeneously broadened lineshapes for both peaks. At room temperature, many vibrational or phonon modes can be populated due to high thermal energy, thus exciton dephasing is dominated by dynamic disorder due to molecular motion in the nanotubes and solvent.

These results are in direct contrast to previous experiments on thin-film, planar Jaggregates that demonstrated dynamic disorder is the primary contributor to exciton dephasing at both room and cryogenic temperatures [155]. However, we compare glass and solution-based nanotubes at 10 K and 295 K, respectively. The dominance of static disorder in contributing to exciton dephasing at 10 K in glass nanotubes may be due to the sucrose-trehalose matrix. The control experiment of comparing the homogeneous and inhomogeneous linewidths of either glass or solution-based nanotubes at room and cryogenic temperatures is not possible due photo-induced degradation and freezing, respectively.

5.4.2 Photo-Induced Degradation and Recovery

Organic semiconductors under high excitation fluences commonly exhibit photo-induced changes due to processes such as exciton-exciton annihilation and charge accumulation [156, 157]. Room temperature measurements of glass J-aggregate nanotubes were not possible due to photodegradation by the ultrafast laser pulses. Photodegradation also occurred at cryogenic temperatures and the lowest possible excitation fluences but was slow enough to not significantly disrupt the 2D measurements.

We studied the effects of photodegradation in glass nanotubes at 10 K by continuously measuring 2D rephasing spectra on the same spot and observing how the spectra changed. Each 2D spectrum took approximately 30 minutes to acquire. The three 2D spectra shown in figure 5.13a represent 90 minutes of continuous irradiation. In these 2D spectra, we observe the three spectral peaks shift to lower wavelength and inhomogeneously broaden



Figure 5.13: Photo-induced degradation and recovery. (a) Irradiation of glass nanotube sample at 10 K for 90 minutes followed by (c) a 90 minute recovery and 30 minutes of reirradiation. (b) Signal intensity decays as a function of irradiation time but returns after a recovery time. (d) Spectral lineshape after 90 minutes of irradiation and recovery is the same as after 60 minutes of irradiation.

as a function of the irradiation time. Figure 5.13b also reveals that the total intensity of the 2D signal decays as a function of the irradiation time. Next, we allowed the sample to recover by turning off the irradiation for 90 minutes, i.e. the original irradiation time. We then performed a final 2D rephasing measurement on the same spot, shown in figure 5.13c, which required an additional 30 minutes of irradiation. In figure 5.13b, we observe there is a recovery in the total 2D signal intensity, while in figure 5.13d, we observe that the spectral lineshape after 90 minutes of recovery is similar to the lineshape after 60 minutes of irradiation. Thus, we observe a recovery in both the signal intensity and spectral lineshape after 90 minutes of non-irradiation.

While the exact mechanism of photodegradation and recovery is still not understood, we have learned that the process is not completely irreversible and thus does not occur due to any significant change in the nanotube structure or dipole-dipole couplings between monomers.

Additionally, our collaborators in the Bawendi group are currently investigating the control of superradiance via optical excitation and the mechanism of photodegradation and recovery in glass nanotubes using linear spectroscopy. Previous studies of a dispersively doped organic dye 1-amino-2-methylanthraquinone (Orange 11) in poly(methyl methacrylate) (PMMA)– analogous to J-aggregate nanotubes in a sucrose-trehalose matrix–also demonstrated selfhealing after photodegradation [158, 159, 160]. The proposed mechanism for photodegradation and recovery was dimerization or aggregation of Orange 11 molecules in the solid state, which required high dyes concentrations. Though this mechanism is unlikely in our system, these experiments set a precedent for the recovery of optical properties via self-healing after photodegradation due to energetic disordering of the system. However, charge accumulation in the glass nanotubes is a likely mechanism for photodegradation and energetic disordering.

5.5 Two-Exciton States in Double-Walled J-Aggregate Nanotubes

Once we characterized the exciton couplings and homogeneous linewidths of single excitons, we measured the two-quantum spectra of both solution-based and glass nanotubes to investigate their two-exciton states.

In the solution-based experiment, isolated nanotubes were flowed through a flow cell to align their long axes to the direction of flow. Co-linear measurements were then performed with the incident laser polarizations parallel to the direction of flow. Under these conditions, only correlations between exciton states polarized parallel to the long axis of the nanotubes (i.e. 599 nm and 598 nm) are interrogated. Figure 5.14a shows the co-linear, two-quantum spectrum of solution-based isolated nanotubes at room temperature. The pure inner and outer wall biexciton states are clearly visible at the two-quantum wavelengths 300 nm and 296 nm, respectively. Both peaks are significantly elongated along the two-quantum axis. The small shoulders to higher wavelength of the pure outer wall biexciton and lower wave-



Figure 5.14: Two-dimensional two-quantum spectra. (a) Two-exciton states in isolated nanotubes at 295 K. The pure inner and outer wall biexcitons are observed at the two-quantum wavelengths 300 nm and 296 nm, respectively. (b) Two-exciton states in glass nanotubes at 10 K. The pure inner and outer wall biexcitons of isolated nanotubes are observed at the two-quantum wavelengths 298 and 293 nm, respectively, and the pure inner wall biexciton of the bundled nanotubes is observed at 296 nm. (c) Isolated nanotube two-quantum spectrum Feynman diagrams (stimulated emission only), where green and blue arrows denote transitions at 599 nm and 589 nm, respectively, and yellow and purple arrows denote transition between $|O_{||}\rangle \leftrightarrow |I_{||}I_{||}\rangle$ and $|I_{||}\rangle \leftrightarrow |O_{||}O_{||}\rangle$, respectively. (d) Bundled nanotube two-quantum spectrum Feynman diagrams.

length of the pure inner wall biexciton suggest mixed inner and outer wall biexcitons may exist, which requires at least weak inter-wall exciton coupling. However, these shoulders span much further than the 298 nm mixed biexciton two-quantum wavelength and extend from 293 nm to 305 nm. Spectral density beyond the mixed biexciton states suggests there may be a transition dipole moment between a single exciton on one wall and a biexciton on the other wall, i.e. $|I\rangle \rightarrow |OO\rangle$ and $|O\rangle \rightarrow |II\rangle$, which also requires inter-wall exciton coupling. Three-dimensional two-quantum measurements could unambiguously determine if the these transition dipole exist by peaks at $(\omega_X, \omega_{YY}, \omega)$. We mention this rather intriguing possibility because experiments using transient grating measurements to perform quantum process tomography on isolated J-aggregate nanotubes in chapter 6 also suggest these transition dipole moments may exist.

In the glass experiment, the samples contained both isolated and bundled nanotubes isotropically oriented and cooled to 10 K. Figure 5.14b shows the co-linear, two-quantum spectrum of glass nanotubes at 10 K. The pure inner and outer wall biexciton states of the isolated nanotubes are clearly visible at 298 nm and 293 nm, respectively. The pure inner wall biexciton state of the bundled nanotubes is also clearly visible at 296 nm. Overall, the two-quantum spectra of the solution-based and glass nanotubes are similar, revealing the same exciton-exciton interactions and suggesting similar excitonic couplings. However, isolated nanotube at room temperature have a large positive biexciton binding energy and fast biexciton dephasing times compared to glass nanotubes at 10 K, which have a nominal biexciton binding energy.

5.6 Conclusions

In this chapter, we used linear absorption as an indicator of the higher-order morphology of double-walled J-aggregate nanotubes. First, we measured the excitonic coupling in solutionbased isolated and bundled nanotubes and in both cases observed strong intra-wall and weak inter-wall exciton coupling. Additionally, two-quantum spectroscopy revealed the presence of pure inner and outer wall biexciton states and a mixed inner and outer wall biexciton state, which again suggested at least weak inter-wall exciton coupling. Investigations into mixed isolated and bundled nanotubes revealed the two morphologies are physically and electronically distinct species in solution. Second, we discovered that J-aggregate nanotubes maintain their self-assembled structure and optical and electronic properties when formed into a glass using a sucrose-trehalose sugar matrix. Temperature-dependent measurements of the homogeneous and inhomogeneous linewidths revealed dynamic and static disorder dominant exciton dephasing at room and cryogenic temperatures, respectively. Investigations into the photo-induced degradation of glass J-aggregate nanotubes at 10 K also demonstrated recovery of their optical properties via self-healing.

Chapter 6

Quantum Process Tomography of Single-Exciton Dynamics in Double-Walled J-Aggregate Nanotubes

In chapter 6, we unambiguously elucidate the single-exciton dynamics in double-walled Jaggregate nanotubes via quantum process tomography using a precisely designed set of narrowband transient grating experiments. For these experiments, J-aggregate nanotubes were synthesized by Dörthe Eisele in the group of Moungi Bawendi at MIT. The transient grating measurements were conducted by Dylan Arias and myself. Quantum process tomography was performed by Joel Yuen-Zhou in the group of Alán Aspuru-Guzik at Harvard University.

First, we will explain why quantum process tomography is a useful tool for elucidating complex quantum dynamics and discuss its general concepts. Second, we will consider the secular dynamics of a two-level system. Third, we will describe our transient grating experiments and explain how quantum dynamics are unambiguously extracted from these data. Finally, we will discuss the quantum dynamics revealed by quantum process tomography for the single-exciton manifold of double-walled J-aggregate nanotubes.

6.1 System-Bath Interactions in the Condensed Phase

The temporal dynamics of quantum systems in the condensed phase are often characterized by complex intra- and inter-system dynamics in and amongst ensemble components, interactions of the system with the environment (or bath), and environmental dynamics [41]. In such cases, the quantum system is generally described by a Hamiltonian that involves separating these interactions using the assumption there is weak coupling between the system and the bath:

$$\hat{H} = \hat{H}_S + \hat{H}_B + \hat{H}_{SB} \tag{6.1}$$

where \hat{H}_S is the Hamiltonian describing the intra- and inter-system dynamics, \hat{H}_B describes the bath dynamics, and \hat{H}_{SB} describes the weakly-coupled system-bath interaction. In the condensed phase, the system dynamics of the ensemble, the bath dynamics of the solvent, and their interactions are generally complex and construction of a Hamiltonian that accurately describes the quantum dynamics of the total system can be difficult. Generally, a Markovian approximation is made which assumes the bath relaxation time is much faster than the relaxation time of the system due to its interaction with the bath, $\tau_S \gg \tau_B$. This approximation requires the system relaxation be stochastic (i.e a random process) but does not reduce the complexity of the system-bath Hamiltonian enough to make it easily tractable. In addition, accurate interpretation of even the simplest time-resolved spectroscopic experiments can be difficult as multiple quantum processes often contribute to single spectral features and separation of these contributions is generally impossible from a single spectroscopic experiment [71, 161, 162].

6.2 Quantum Process Tomography

A new approach for understanding the temporal dynamics of complex quantum systems comes from the field of quantum information processing called quantum process tomography (QPT) [163, 164, 165, 72]. The goal of QPT is to completely elucidate the time-dependent density matrix of a system $\rho(T)$ by using a precisely designed set of spectroscopic experiments [166]. In QPT, the quantum dynamics of the system are characterized by a timedependent process matrix $\chi(T)$ [different from the susceptibility in chapter 3] that describes the evolution of an initially prepared quantum state at time zero $\rho(0)$ to a final state at some time T later $\rho(T)$. In this paradigm, the process matrix characterizes all the system and bath dynamics and interactions and acts on the system as a linear transformation between its initial and final states:

$$\rho(T) = \chi(T)\rho(0) \tag{6.2}$$

In spectroscopic experiments, an initial state $\rho(0)$ is prepared by a set of excitation pulses, $\chi(T)$ propagates the density matrix during time T, and the final state $\rho(T)$ is interrogated by a set of detection pulses. When expanded into the basis states of the system, equation 6.2 takes the form:

$$\rho_{cd}\left(T\right) = \sum_{ab} \chi_{cdab}\left(T\right) \rho_{ab}\left(0\right) \tag{6.3}$$

where the time-dependent process matrix element $\chi_{cdab}(T)$ describes the conversion of the initially prepared density matrix element $\rho_{ab}(0)$ into the final density matrix element $\rho_{cd}(T)$ at some time T. Thus, process matrix elements of the form χ_{aaaa} describe the decay of a population (i.e. diagonal density matrix element) state a, χ_{abab} describe the decay of a coherence (i.e. off-diagonal density matrix element) between states a and b, and χ_{bbaa} describe the population transfer from state a to state b. The process matrix elements χ_{bcaa} , χ_{ccab} , and χ_{cdab} also describe the non-secular processes of population-to-coherence transfer, coherence-to-population transfer, and coherence transfer, respectively.

To perform QPT, a precisely designed set of spectroscopic experiments is used to selectively prepare every possible initial state and measure their conversion into every possible final state [167, 168, 169]. In this procedure, the construction of an accurate Hamiltonian for the total system is unnecessary because all the quantum dynamics are extracted directly from the spectroscopic measurements. However, a basic model for the system and an accurate understanding of the linear absorption spectrum is necessary to interpret the results of QPT. In this chapter, we use QPT to study the single-exciton dynamics of double-walled J-aggregate nanotubes, which were extensively studied and well characterized by linear and two-dimensional spectroscopy in chapter 5. While QPT has been previously used to study the quantum dynamics of spins, qubits, ions, and optical lattices, our experiments are the first performed on a molecular or supramolecular system, in the condensed phase, or at room temperature [170, 171, 172, 173, 174, 175].

6.3 Secular Dynamics of a Two-Level System

To acquire a basic understanding of the quantum dynamics underlying a complex condensed phase system, we will consider the case of a two-level excited state system and explore its secular dynamics: population decay and transfer and coherence decay (i.e. dephasing). In our model system, we define the excited states a and b with energies $\hbar\omega_a$ and $\hbar\omega_b$, respectively, where $\omega_a > \omega_b$ and $\omega_{ab} = \omega_a - \omega_b$ is the difference frequency between the states.

$$E = \hbar \begin{bmatrix} \omega_b & \omega_{ba} \\ \omega_{ab} & \omega_a \end{bmatrix}$$
(6.4)

The density matrix of the excited states thus takes the form:

$$\rho = \begin{bmatrix} |b\rangle \langle b| & |b\rangle \langle a| \\ |a\rangle \langle b| & |a\rangle \langle a| \end{bmatrix}$$
(6.5)

where the diagonal matrix elements $|x\rangle \langle x|$ describe excited state populations and the offdiagonal matrix elements $|x\rangle \langle y|$ describe coherences between the excited states. We also define a rate matrix R characterizing the time constants for each excited state population and coherence decay and transfer process:

$$R = \begin{bmatrix} R_{bbbb} & R_{aabb} & R_{abbb} & R_{babb} \\ R_{bbaa} & R_{aaaa} & R_{abaa} & R_{baaa} \\ R_{bbab} & R_{aaab} & R_{abab} & R_{baab} \\ R_{bbba} & R_{aaba} & R_{abba} & R_{baba} \end{bmatrix}$$
(6.6)

Now, the temporal dynamics of the excited populations and coherences can be described using normal chemical kinetics. The population kinetics take the form:

$$\dot{\rho}_{aa}\left(t\right) = R_{aabb}\rho_{bb}\left(t\right) - R_{bbaa}\rho_{aa}\left(t\right) \tag{6.7}$$

$$\dot{\rho}_{bb}(t) = R_{bbaa}\rho_{aa}(t) - R_{aabb}\rho_{bb}(t)$$
(6.8)

where the change in each population is described by the transfer into (i.e. positive) and out of (i.e. negative) each state. If we assume uphill population transfer does not occur (i.e. $R_{aabb} = 0$), the time derivatives of the excited state populations reduce to:

$$\dot{\rho}_{aa}\left(t\right) = -R_{bbaa}\rho_{aa}\left(t\right) \tag{6.9}$$

$$\dot{\rho}_{bb}\left(t\right) = R_{bbaa}\rho_{aa}\left(t\right) \tag{6.10}$$

In this case, the population dynamics are completely defined by the downhill population transfer rate from excited state a to b (R_{bbaa}). Integrating the differential equations using the initial condition that the excited state population begins in state a (i.e. $\rho_{aa}(0) = 1$ and $\rho_{bb}(0) = 0$), the kinetics for the excited state populations can be explicitly defined.

$$\rho_{aa}\left(t\right) = \exp\left[-R_{bbaa}t\right] \tag{6.11}$$

$$\rho_{bb}(t) = 1 - \exp\left[-R_{bbaa}t\right] \tag{6.12}$$

Under these initial conditions, the population in excited state a starts at 1 and decays

exponentially, while the population in excited state b starts at 0 and simultaneously rises exponentially on the timescale $1/R_{bbaa}$ due to downhill population transfer. We implicitly assume the decay rate between the excited and ground states is much slower than the transfer rate between excited states and thus, in a short time approximation, decay to the ground state is does not occur.

Similar to the temporal dynamics of the populations, the kinetics of the coherences can be written down and take the form:

$$\dot{\rho}_{ab}(t) = -(i\omega_{ab} + R_{abab})\rho_{ab}(t)$$
(6.13)

$$\dot{\rho}_{ba}(t) = -(i\omega_{ba} + R_{baba})\rho_{ba}(t) \tag{6.14}$$

We note $\dot{\rho}_{ab}(t) = \dot{\rho}_{ba}^{*}(t)$ because the density matrix is Hermitian (i.e. self-adjoint). Integrating the differential equation using the initial condition that the amplitude of the coherences is unity (i.e. $\rho_{ab}(0) = \rho_{ba}(0) = 1$), the kinetics for the excited state coherences can be explicitly defined.

$$\rho_{ab}\left(t\right) = \rho_{ba}^{*}\left(t\right) = \exp\left[-\left(i\omega_{ab} + R_{abab}\right)t\right]$$
(6.15)

Under these initial conditions, the excited state coherences start at 1, oscillates at the difference frequency $\omega_{ab} = -\omega_{ba}$, and dephases exponentially on the timescale $1/R_{abab} = 1/R_{baba}$.

6.4 Narrowband Transient Grating Experiments

With a basic understanding of the secular dynamics of a two-level excited state system, we can now describe the transient grating measurements used for QPT. To elucidate the process matrix of a quantum system, the spectroscopic experiments must selectively excite every possible quantum state (both populations and coherences) and selectively detect their evolution into every other quantum state. In our experimental design, we use transient grating spectroscopy to achieve the excitation and detection selectively required by QPT. To

describe our experimental measurements, we will consider the case of two optically accessible excited states a and b with non-overlapping absorption spectra. First, two narrowband pump pulses selectively excite one of the four possible quantum states: the $|b\rangle \langle b|$ and $|a\rangle \langle a|$ excited state populations and the $|b\rangle \langle a|$ and $|a\rangle \langle b|$ excited state coherences. The first pump pulse (wavevector $-\vec{k_1}$) acts on the ket and the second pump pulse $(\vec{k_2})$ acts on the bra to prepare the desired quantum state. The crossed pump pulses create an excited state grating off which the third narrowband probe pulse (\vec{k}_3) scatters after a delay time T. The scattered probe pulse is heterodyne detected with a fourth broadband pulse $(\vec{k}_4 = -\vec{k}_1 + \vec{k}_2 + \vec{k}_3)$ by spectral interferometry using a spectrometer. The central frequencies of the narrowband pump pulses determine the selectively prepared initial state, while the central frequency of the narrowband probe pulse determines the selectively detected final state. The fourth pulse does not contribute to the scattered probe signal but acts as a broadband reference to detect both elastic and inelastic probe pulse scattering and does not effect the selectively of the spectroscopic measurements. In this experimental design, eight unique measurements are required to implement QPT. We label the experiments xyz, where the initial state $|x\rangle\langle y|$ is selectively prepared and the state $|z\rangle$ is selectively detected and $x, y, z \in a, b$. Even with the excitation and detection selectively of the transient grating measurements, a single spectroscopic signal can contain contributions from multiple process matrix elements. However, the set of transient grating measurements described can be used to extract the process matrix by implementing QPT.

We use QPT to study the single-exciton dynamics of double-walled J-aggregate nanotubes, illustrated in figure 6.1a [146, 147, 176]. Our measurements are the first realization of QPT in a molecular or supramolecular system, in the condensed phase, or at room temperature. In our experiments, the nanotubes are flowed through a cell and align along the direction of flow. The excitation and detection pulses are polarized parallel to the flow direction, so only the primary inner and outer wall transitions at 16,700 and 17,000 cm⁻¹, respectively, are interrogated. The transient grating measurements thus elucidate the dy-



Figure 6.1: Quantum process tomography of double-walled J-aggregate nanotubes. (a) J-aggregate nanotube consisting of an inner and outer wall. (b) Linear absorption of states polarized parallel to the long axis of the nanotube (red) and the narrowband pulses resonant with the inner and outer wall absorption (green and blue, respectively) used in the narrowband transient grating measurements. (c) Energy level diagram of the J-aggregate nanotubes including the ground, one-exciton, and two-exciton manifolds. E_x is the energy and $\mu_{g,x}$ the transition dipole moment between the ground $|g\rangle$ and one-exciton state $|x\rangle$. The two-exciton states $|yz\rangle$ are characterized by an additional biexciton binding energy ΔE_{yz} and the one-to two-exciton transition dipole moment $\mu_{x,yz}$.

namics of a two-level, one-exciton excited state manifold analogous to the quantum system described above. Figure 6.1b shows the linear absorption of the exciton states polarized along the long axis of the nanotube and the narrowband pulses used for excitation and detection at the central frequencies of the inner and outer wall exciton resonances. The narrowband pulses allow selective excitation and detection while maintaining a fairly high time resolution of approximately 100 fs. Two-dimensional phase and amplitude shaping of broadband pulses is used to generate and compress the frequency-selective narrowband pulses, which each had a pulse energy of approximately 0.5 nJ.

To fully elucidate the quantum dynamics of the single-exciton manifold, the ground and two-exciton manifolds must also be considered. The energy level diagram in figure 6.1c illustrates the accessible exciton states in our transient grating measurements along with their energies and transition dipole moments. These states include two pure and one mixed biexciton states $|xx\rangle$ and $|xy\rangle$, respectively, which are characterized by biexciton binding energies ΔE_{xy} . The biexciton binding energies are small compared to the full width at half maximum of the linear absorption peaks, thus signals emitting from the one- and two-exciton states are not spectrally separated and overlap in the transient grating measurements.

We performed eight transient grating measurements labelled: *OOO*, *OOI*, *IIO*, *III*, *OIO*, *OII*, *IOO*, *IOI* (as described above). The quantum states of interest are described by the single-exciton density matrix:

$$\rho_{SE} = \begin{bmatrix} |I\rangle \langle I| & |I\rangle \langle O| \\ |O\rangle \langle I| & |O\rangle \langle O| \end{bmatrix}$$
(6.16)

Each quantum state is selectively prepared by the two narrowband pump pulses and their emission is selectively interrogated by the narrowband probe pulse at the central frequency of either the inner or outer wall absorption, I or O, respectively. The raw transient grating data from the eight experiments are plotted in figure 6.2. As mentioned previously, the signals from the eight transient grating measurements do not inherently separate the elements of the process matrix. By elucidating all the possible Feynman diagrams for each measurement and allowing transfer from the initial quantum state to all other quantum states, we can determine which process matrix elements contribute to the spectroscopic signals in our experiments. Figure 6.3 shows Feynman diagrams for the preparation of each quantum state and the emission from each final state after delay time T-which is dependent on the central frequency of the third narrowband pulse. The full set of Feynman diagrams for our QPT experiments is provided in Appendix A. In table 6.1, we summarize the information provided by the full set of Feynman diagrams.

The process matrix elements enumerated in table 6.1 take the form χ_{cdab} , where the quantum state $|a\rangle \langle b|$ evolves into state $|c\rangle \langle d|$ during the experimental delay time T. The process matrix elements include all secular processes: population decay χ_{aaaa} , population transfer χ_{bbaa} , and coherence decay χ_{abab} ; and non-secular processes: population-to-coherence transfer χ_{bcaa} , coherence-to-population transfer χ_{ccab} , and coherence transfer χ_{cdab} , where $a, b, c, d \in I, O$. From table 6.1, we observe that secular and non-secular processes emit at different frequencies in each of the eight transient grating measurements. From this



narrowband probe pulse centered at the central frequency of state z selectively detecting the state $|z\rangle$, where x, y, $z \in I, O$. centered at the central frequencies of states x and y, respectively, selectively generating the initial state $|x\rangle \langle y|$ and the third Figure 6.2: Transient grating measurements. Measurement xyz is characterized by the first and second narrowband pump pulses



Figure 6.3: Transient grating Feynman diagrams for quantum process tomography. Bottom Row: Two narrowband pulse preparation of the initial states $|O\rangle \langle O|$, $|I\rangle \langle I|$, $|O\rangle \langle I|$, and $|I\rangle \langle O|$ followed by system evolution during the waiting time T. Top Rows: Third narrowband pulse-dependent emission of the final states $|O\rangle \langle O|$, $|I\rangle \langle I|$, $|O\rangle \langle I|$, and $|I\rangle \langle O|$ including stimulated emission and excited state absorption pathways, where ΔE_{xy} denotes the binding energy of the $|xy\rangle$ biexciton. Green and blue arrows indicate interactions with energies E_I and E_O , respectively. Light green and light blue arrows indicate interactions with energies shifted by the biexciton binding energies $E_I - \Delta E$ and $E_O - \Delta E$, respectively. Yellow and purple arrows indicate interactions with energies $(2E_I - \Delta E_{II}) - E_O$ and $(2E_O - \Delta E_{OO}) - E_I$, respectively.

						-		
Emission Frequency	000	OOI	IIO	III	OIO	OII	IOO	IOI
1. $2E_O - \Delta E_{OO} - E_I$	<i>ΧΙΟΟΟ</i>		χιοιι		χιοοι		χιοιο	
2. E_O	χοοοο	<i>X1000</i>	χοοιι	χ_{IOII}	χοοοι	χ_{IOOI}	χοοιο	χιοιο
3. $E_O - \Delta E_{IO}$	<i>ΧΙΙΟΟ</i>	<i>X1000</i>	χ_{IIII}	χ_{IOII}	χ_{IIOI}	χ_{IOOI}	χ_{IIIO}	χιοιο
4. $E_O - \Delta E_{OO}$	χοοοο		χοοιι		χοοοι		χοοιο	
5. E_I	χοιοο	χίιοο	χοιιι	χ_{IIII}	χοιοι	χ_{IIOI}	χοιιο	χ_{IIIO}
6. $E_I - \Delta E_{IO}$	χοιοο	χοοοο	χοιιι	χοοιι	χοιοι	χοοοι	χοιιο	χοοιο
7. $E_I - \Delta E_{II}$		χιιοο		χ_{IIII}		χ_{IIOI}		χ_{IIIO}
8. $2E_I - \Delta E_{II} - E_O$		χοιοο		χοιιι		χοιοι		χοπο

Process Matrix Contributions to Transient Grating Measurements

Table 6.1: Process matrix contributions to transient grating measurements. The process matrix elements χ_{cdab} are categorized by their emission energy (1-8) in each transient grating measurement xyz. Measurement xyz selectively generates the initial state $|x\rangle \langle y|$ and selectively detects state $|z\rangle$, where $x, y, z \in I$, O. Red process matrix elements denote nonsecular processes: population-to-coherence transfer, coherence-to-population transfer, and coherence transfer.

observation, we can immediately determine from the raw transient grating data that nonsecular processes are not observed in our experiments, which will be confirmed by the QPT procedure. We can also directly observe from the oscillations in the OIO, OII, IOO, and IOI experiments that coherences between excitons on the inner and outer walls exist and thus the states must be inherently coupled.

We can write down time-dependent equations for each transient grating measurement with contributions from secular process matrix elements only, assuming no uphill population transfer (i.e. $\chi_{OOII} = 0$):

$$OOO(T, t) = \chi_{OOOO}(T) \left[|\mu_{g,O}|^4 \exp\left[- (i\omega_O + \gamma_{g,O}) t \right] - |\mu_{g,O}|^2 |\mu_{O,OO}|^2 \exp\left[- (i(\omega_O - \Delta\omega_{OO}) + \gamma_{O,OO}) t \right] \right] - \chi_{IIOO}(T) |\mu_{g,O}|^2 |\mu_{I,IO}|^2 \exp\left[- (i(\omega_O - \Delta\omega_{IO}) + \gamma_{I,IO}) t \right]$$
(6.17)
$$OOI(T, t) = -\chi_{OOOO}(T) |\mu_{g,O}|^2 |\mu_{O,IO}|^2 \exp\left[- (i\omega_I - \Delta\omega_{IO} + \gamma_{O,IO}) t \right] + \chi_{IIOO}(T) \left[|\mu_{g,O}|^2 |\mu_{g,I}|^2 \exp\left[- (i\omega_I + \gamma_{g,I}) t \right] - |\mu_{g,O}|^2 |\mu_{I,II}|^2 \exp\left[- (i(\omega_I - \Delta\omega_{II}) + \gamma_{I,II}) t \right] \right]$$
(6.18)

$$IIO(T,t) = -\chi_{IIII}(T) |\mu_{g,I}|^2 |\mu_{I,IO}|^2 \exp\left[-\left(i\left(\omega_O - \Delta\omega_{IO}\right) + \gamma_{I,IO}\right)t\right]$$
(6.19)

$$III(T,t) = \chi_{IIII}(T) \left[|\mu_{g,I}|^4 \exp\left[-(i\omega_I + \gamma_{g,I}) t \right] - |\mu_I|^2 |\mu_{I,II}|^2 \exp\left[-(i(\omega_I - \Delta\omega_{II}) + \gamma_{I,II}) t \right] \right]$$
(6.20)

$$OIO(T, t) = \chi_{OIOI}(T) [|\mu_{g,O}\mu_{g,I}\mu_{g,O}\mu_{g,I}| \exp[-(i\omega_{I} + \gamma_{g,I})t] - |\mu_{g,O}\mu_{g,I}\mu_{I,IO}\mu_{O,IO}| \exp[-(i(\omega_{I} - \Delta\omega_{IO}) + \gamma_{O,IO})t]]$$
(6.21)

$$OII(T,t) = -\chi_{OIOI}(T) |\mu_{g,O}\mu_{g,I}\mu_{I,II}\mu_{O,II}| \times \exp\left[-\left(i\left(2\omega_{I} - \Delta\omega_{II} - \omega_{O}\right) + \gamma_{O,II}\right)t\right]$$
(6.22)

$$IOO(T,t) = -\chi_{IOIO}(T) |\mu_{g,I}\mu_{g,O}\mu_{O,OO}\mu_{I,OO}| \times \exp\left[-\left(i\left(2\omega_O - \Delta\omega_{OO} - \omega_I\right) + \gamma_{I,OO}\right)t\right]$$
(6.23)

$$IOI(T,t) = \chi_{IOIO}(T) [|\mu_{g,I}\mu_{g,O}\mu_{g,I}\mu_{g,O}| \exp[-(i\omega_O + \gamma_{g,O})t] - |\mu_{g,I}\mu_{g,O}\mu_{O,IO}\mu_{I,IO}| \exp[-(i(\omega_O - \Delta\omega_{IO}) + \gamma_{I,IO})t]]$$
(6.24)

The initial intensity for each signal contribution is determined by the product of the relevant transition dipole moments between states j and k, $\prod^4 |\mu_{j,k}|$, and exist between but not within the ground state, one-, and two-exciton manifolds. The delay time dynamics are completely governed by the process matrix elements $\chi_{cdab}(T)$. The signal is determined by the radiating coherence during the emission time t, $\exp[-(i\omega_j + \gamma_{j,k})t]$, and characterized by the exciton energies $E_j = \hbar \omega_j$ and biexciton binding energies $\Delta E_{jk} = \hbar \Delta \omega_{jk}$ and the dephasing rates $\gamma_{j,k}$ between states j and k. The total signal for each transient grating measurement is the sum over all the contributing process matrix elements.

With the eight transient grating measurements and analytical expressions for each of the spectroscopic signals, QPT can invert the data to separate and unambiguously elucidate the dynamics of each process matrix element.

6.5 Signal Inversion and Process Matrix Extraction

The time-dependent signals for the transient grating measurements in equations 6.17-6.24 are composed of linear combinations of the process matrix elements and are scaled by the relevant transition dipole moments. Thus, the transient grating signals must be inverted to determine the process matrix elements.

Prior to implementing the QPT procedure, the transient grating signals were normalized to account for three variable experimental parameters. First, the signals were normalized by the relevant pulse energies for each of the three narrowband pulses. Second, the nanotubes degrade slightly over time due to pumping through the flow cell and optical excitation, so the signals were normalized by the optical density of the nanotubes by taking a linear absorption spectrum after each experiment. Finally, the transient grating signals were corrected for pulse intensity rolloff (described in chapter 4) due to time delays applied via temporal pulse shaping.

To implement the QPT procedure and extract the process matrix elements, first the transient grating signals are integrated over the emission frequency, $\int d\omega XYZ(T,\omega) = XYZ(T,t=0)$, yielding the time-domain signal at t=0 according to Fourier transform relationships and eliminating the exponential signal terms in equations 6.17-6.24. Second, the signals with initial excited state populations (i.e. OOY and IIY) are corrected by subtracting the signal at T=0, $XXY(T-T_0) = XXY(T) - XXY(0)$, to remove interfering spectroscopic signals due to ground state bleach that do not contribute to the single-exciton dynamics. Third, the transient grating signals are inverted to generate expressions for the process matrix elements. Inversion utilizes semi-definite programming to ensure the process matrix is a linear transformation that maps physical density matrices to other physical density matrices that are Hermitian, trace preserving, and possess positive populations [177, 178].

From equations 6.19-6.24, we observe that the transient grating experiments IIY, OIY, and IOY are dependent on a single process matrix element χ_{IIII} , χ_{OIOI} , and χ_{IOIO} , respectively. Thus, inversion of these process matrix elements are straightforward. From equations 6.17 and 6.18, we observe the OOY experiments are coupled and depend on the process matrix elements χ_{OOOO} and χ_{IIOO} . Thus, the process matrix elements χ_{OOOO} and χ_{IIOO} must be solved simultaneously, which is also straightforward but slightly more cumbersome algebraically.

The process matrix elements extracted from equations 6.17-6.24 are:

$$\chi_{OOOO}(T) = 1 + \frac{\left(\frac{C_3^{OOO} - C_1^{OOI}}{C_1^{OOO} - C_2^{OOO}}\right)^{OOO(T - T_0) + C_3^{OOO} OOI(T - T_0)}}{\left(C_2^{OOI} - C_3^{OOI}\right) - \left(\frac{C_3^{OOO} - C_1^{OOI}}{C_1^{OOO} - C_2^{OOO}}\right)} (6.25)$$

$$\chi_{IIOO}(T) = \frac{OOI(T - T_0) + \left(\frac{C_1^{OOI}}{C_1^{OOO} - C_2^{OOO}}\right) OOO(T - T_0)}{(C_2^{OOI} - C_3^{OOI}) - \left(\frac{C_3^{OOO} - C_1^{OOI}}{C_2^{OOO} - C_2^{OOO}}\right)}$$
(6.26)

$$\chi_{IIII}(T) = 1 + \frac{III(T - T_0)}{(C_1^{III} - C_2^{III})} = 1 - \frac{IIO(T - T_0)}{C_1^{IIO}}$$
(6.27)

$$\chi_{OIOI}(T) = = \frac{OIO(T)}{(C_1^{OIO} - C_2^{OIO})} = \frac{OII(T)}{C_1^{OII}}$$
(6.28)

$$\chi_{IOIO}(T) = \frac{IOI(T)}{(C_1^{IOI} - C_2^{IOI})} = \frac{IOO(T)}{C_1^{IOO}}$$
(6.29)

The coefficients c_j^{XYZ} are time-dependent and defined in equations 6.17-6.24 such that $XYZ(T) = \sum_j c_j^{XYZ}(t) \chi_j(T)$. These coefficients are products of the transition dipole moments and complex exponentials involving the transition energies and phenomenological dephasing rates. Thus, a model for the quantum system is required. A comprehensive understanding of the linear absorption spectrum (described in chapter 5 for double-walled J-aggregate nanotubes) is sufficient to characterize the system and determine the necessary parameters to calculate the time-dependent coefficients c_j^{XYZ} . However, extraction of the transition dipole moments from the transient grating measurements, which is robust and easy to achieve, leads to more accurate normalization of the signal contributions to compute the process matrix elements.

Because only secular dynamics and downhill energy transfer occur and population re-

laxation to the ground state is slow, the expressions for the process matrix elements in equations 6.25-6.29 are significantly reduced in complexity. The process matrix elements χ_{IIII} , χ_{OIOI} , and χ_{IOIO} can be singularly described by one of two transient grating measurements that provide redundant information, while the process matrix elements χ_{OOOO} and χ_{IIOO} are slightly more complex and possess contributions from two transient grating measurements. To reduce the complexity of the process matrix elements χ_{OOOO} and χ_{IIOO} , we note the transient grating signal amplitude for experiment *IIO* is nearly zero compared to other experiments. From equation 6.19, we observe the *IIO* transient grating measurement is compromised solely of an excited state absorption pathway requiring the mixed biexciton state $|IO\rangle$, which we will assume does not contribute substantially to the transient grating signals. Thus, the coefficients c_3^{OOO} and c_1^{OOI} are zero and equations 6.25 and 6.26 reduce to:

$$\chi_{OOOO}(T) = 1 + \frac{OOO(T - T_0)}{(C_1^{OOO} - C_2^{OOO})}$$
(6.30)

$$\chi_{IIOO}(T) = \frac{OOI(T - T_0)}{(C_2^{OOI} - C_3^{OOI})}$$
(6.31)

If non-secular dynamics and uphill energy transfer occurred and the mixed biexciton state $|IO\rangle$ contributed, the process matrix elements would depend on many more transient grating measurements and inversion would be more cumbersome. Table 6.2 summarizes the transient grating signal contributions at a given emission frequency (where α is the emission frequency number defined in table 6.1) for each process matrix element if all quantum processes occurred.

χοοοο	χιιοο	χοιοο	<i>Χιοοο</i>	χοοιι	ΧΙΙΙΙ	χοιιι	χιοιι
OOO(2)	OOO(3)	OOO(5)	OOO(1)	IIO(2)	IIO(3)	IIO(5)	IIO(1)
$OOO\left(4\right)$	OOI(5)	OOO(6)	OOI(2)	IIO(4)	III(5)	IIO(6)	III(2)
$OOI\left(6 ight)$	OOI(7)	$OOI\left(8 ight)$	OOI(3)	III(6)	III(7)	III(8)	$III\left(3 ight)$
χοοοι	χ_{IIOI}	χοιοι	χιοοι	χοοιο	χ_{IIIO}	χοιιο	χιοιο
OIO(2)	OIO(3)	OIO(5)	OIO(1)	IOO(2)	IOO(3)	IOO(5)	IOO(1)
$OIO\left(4 ight)$	OII(5)	$OIO\left(6 ight)$	OII(2)	IOO(4)	IOI(5)	IOO(6)	IOI(2)
OII(6)	OII(7)	OII(8)	OII(3)	IOI(6)	IOI(7)	IOI(8)	IOI(3)

Transient Grating Signal Contributions to Process Matrix Elements

Table 6.2: Transient grating signal contributions to process matrix elements. $xyz(\alpha)$ denotes the transient grating signal of experiment xyz at emission frequency number α as defined in table 6.1. Measurement xyz selectively generates the initial state $|x\rangle \langle y|$ and selectively detects state $|z\rangle$, where $x, y, z \in I$, O. Red process matrix elements denote non-secular processes: population-to-coherence transfer, coherence-to-population transfer, and coherence transfer.

6.6 Quantum Dynamics of Double-Walled J-Aggregate Nanotubes

From the raw transient grating data, we were able to directly observe that non-secular processes: population-to-coherence transfer (i.e. χ_{OIOO} , χ_{IOOO} , χ_{OIII} , and χ_{IOII}), coherence-topopulation transfer (i.e. χ_{OOOI} , χ_{IIOI} , χ_{OOIO} , and χ_{IIIO}), and coherence transfer (i.e. χ_{IOOI} and χ_{OIIO}), are negligible in our system. Thus, only the process matrix elements for secular processes: population decay (i.e. χ_{OOOO} and χ_{IIII}), population transfer (i.e. χ_{IIOO} and χ_{OOII}), and coherence decay (i.e. χ_{OIOI} and χ_{IOIO}), are non-zero. All the time-dependent process matrix elements inverted from the transient grating measurements are plotted in figure 6.4.

Indeed, we observe the non-secular process matrix elements including population-tocoherence transfer, coherence-to-population transfer, and coherence transfer are all zero. The exciton population in the outer wall decays and transfers completely to the inner wall within 300 fs, observed from χ_{OOOO} and χ_{IIOO} , respectively. However, the exciton population in the inner wall does not decay, $\chi_{IIII}(T) = 1$, due to the slow relaxation time to the ground state. Additionally, inner wall excitons do not transfer uphill to the outer wall, $\chi_{OOII}(T) = 0$,



Figure 6.4: Time-dependent process matrix elements. Including secular processes: population decay (χ_{OOOO} and χ_{IIII}), population transfer (χ_{IIOO} and χ_{OOII}), and coherence decay (χ_{OIOI} and χ_{IOIO}) and non-secular processes: population-to-coherence transfer (χ_{OIOO} , χ_{IOOO} , χ_{OIII} , and χ_{IOII}), coherence-to-population transfer (χ_{OOOI} , χ_{IIOI} , χ_{OOIO} , and χ_{IIIO}), and coherence transfer (χ_{IOOI} and χ_{OIIO}). \Re and \Im denote the real and imaginary components of a given process matrix element.

ecay	Parameters	$2\pi/\omega_{OI} = 70 \pm 0.5 fs$ $ au_{OI} = 155 \pm 10 fs$	$\beta_{OI} = 2.6 \pm 0.5$	nsfer	Parameters	I		ion Transfer	Parameters		·	,
Coherence De	Fit	$\exp\left[-i\omega_{OI}T\right] \times \\ \exp\left[-\left(T/\tau_{O}\right)\right] \partial_{OOI}^{2} d$	$\left[10 d \left(10 r \right) + 10 d \right] d v a$	Coherence Tra	Fit	0		Coherence-to-Populat	Fit	0	0	
	×	$\chi_{OIOI} = \chi_{\chi^*}$	XIOIO		X	$\chi_{IOOI} = \chi_{\star}$	χ_{OIIO}		X	$\chi_{IIIO} = \chi_{IIIO}^*$	$\begin{array}{c} \chi_{0000} = \\ \chi_{0010}^* \end{array}$	
Population Decay	Parameters	$\tau_{OO} = 212 \pm 8fs$ $\beta_{OO} = 3.3 \pm 0.4$	ı	Population Transfer	Parameters	$\tau_{OO} = 212 \pm 8fs$ $\beta_{OO} = 3.3 \pm 0.4$	I	opulation-to-Coherence Transfer	Parameters	I	ı	•
	Fit	$\exp\left[-\left(T/\tau_{OO}\right)\beta_{OO}\right]$	1		Fit	$1 - \exp\left[-\left(T/\tau_{OO}\right)\beta_{OO}\right]$	0		Fit	0	0	
	X	χοοοο	XIIII		X	οοπχ	Χοοιι	P.	X	$\chi_{0000} = \chi_{0100}^*$	$\chi^{NIIII}_{OIII} = \chi^*_{OIII}$; ; ; ;

χ_{IIII}),	ation-to-	IO, and	escale of	
200 and	popula	and χ_{II}	ion time	
ay (χ_{OG}	D Cesses:	$\chi_{OOIO},$	relaxat	
ion dec	ılar pro	χ_{IIOI}, χ	icy and	
opulat	non-sec	<i>Χοοοι</i> ,	frequer	amics.
esses: 1) and 1	nsfer (z)	illation	ne dyna
ar proc	$1 \chi_{IOIC}$	tion tra	the osc	ntial tir
g secula	or and	popula	denote	expone
Inding	ay (χ_O)	nce-to-	nd τ_{xy}	or the a
ers. Ir	ice dec	cohere	S ω_{xy} a	meter f
aramet	coherer	$\chi_{IOII}),$	rameter	ıg para
s and p), and	$_{II}$, and	The pa	tretchi
ent fits	χοοιι	ю, Хоі.	o_{IIO} .	$_{vy}$ the s
x elem	o and	$, \chi_{IOO}$	and χ	and β_a
; matri	r (χ_{IIO}	(χ_{OIOC})	(χ_{IOOI})	$ x\rangle \langle y $
Process	transfe	ransfer	ansfer	n state
9 6.3:	lation	rence ti	rence ti	luantur
Tablé	ndod	cohei	coheı	the q

because the 430 cm⁻¹ frequency difference is large compared to k_BT at 298 K, which is 200 cm⁻¹. The real and imaginary components of χ_{OIOI} reveal coherences between inner and outer wall excitons oscillating at the 430 cm⁻¹ difference frequency and lasting approximately 300 fs, signifying weak coupling between the states. Coupling between inner and outer wall excitons is also consistent with the small but measurable crosspeaks between the 16,700 and 17,000 cm⁻¹ inner and outer wall peaks in the two-dimensional spectra presented in chapter 5.

The data in figure 6.4 can be fit to the models derived in equations 6.11-6.15 and quantitative parameters can be determined. The fits, parameters, and error bars for each process matrix element is summarized in table 6.3. The experimental data are fit to stretched rather than standard exponentials and are characterized by the stretching parameter β_x , which we speculate is due to exponential kinetics that are embedded in Gaussian disorder.

6.7 Conclusions

Previous pump-probe measurements have attempted to characterize excited state population transfer and decay but were unable to unambiguously separate the inner and outer wall exciton dynamics or determine the role of biexciton states [145, 179, 151, 180]. Similarly, the two-dimensional spectra in chapter 5 could not completely separate the process matrix elements by spreading the signal over a second frequency axis as the crosspeaks include multiple contributions that evolve during the population time [47, 150, 48, 181, 176]. However, our narrowband transient grating measurements and the implementation of QPT unequivocally elucidate the single-exciton dynamics of double-walled J-aggregate nanotubes. First, we observed that non-secular processes and uphill energy transfer do not occur in our system. Second, we observed downhill energy transfer from the outer to inner wall occurred on a timescale of 212 ± 8 fs. Third, we observed weak coupling between inner and outer wall excitons characterized by dephasing on a timescale of 155 ± 10 fs. These experiments
are the first realization of QPT in a molecular or supramolecular system, in the condensed phase, or at room temperature and are the first experiments to unambiguously characterize the quantum dynamics of excitons in double-walled J-aggregate nanotubes.

Chapter 7

Two-Body Exciton Interactions in Cuprous Oxide

In chapter 7, we describe the first successful measurements of bi-orthoexciton coherences in cuprous oxide using two-dimensional two-quantum spectroscopy. In cuprous oxide, biexcitons are believed to drive inelastic exciton scattering but have not been detected because of strict optical selection rules that forbid radiative biexciton decay. We directly observe bi-orthoexcitons in cuprous oxide and estimate a 13 meV bi-orthoexciton binding energy, which is in the range of previous theoretical predictions, and a coherence lifetime around 70 fs. The creation or annihilation of a phonon accompanying an electronic transition also substantially increases the magnitude of the detected bi-orthoexciton coherence.

For these experiments, the cuprous oxide crystal was grown by Laszlo Frazer in the group of John Ketterson and Kelvin Chang in the group of Kenneth Poeppelmeier at Northwestern University. The two-dimensional spectroscopy measurements were conducted by Yoseob Yoon and myself.

7.1 Cuprous Oxide

Excitons in cuprous oxide (Cu₂O) are bound electron-hole pairs. The O_h symmetry of Cu₂O crystals and the direct gap d to s band transitions, which form the basis of the lowest lying "yellow" exciton series, give rise to a near-Rydberg structure [182]. However, the 1s states deviate from the Rydberg pattern [183]. The high symmetry of the 1s paraexciton singlet makes it optically inaccessible to all orders in the multipole expansion [184]. The 12 meV higher energy 1s orthoexciton triplet state has single-photon transitions that are dipole-forbidden but quadrupole-allowed through an exceptionally narrow polariton resonance [185, 186]. The optical selection rules are so strict that, despite the direct gap, the strongest transitions require the participation of a phonon [186, 187, 188, 189].

Bound pairs of 1s yellow orthoexcitons (i.e. bi-orthoexcitons) are expected to be forbidden from coupling to light to all orders, similar to paraexcitons [190]. Previously, luminescence data were interpreted as evidence of a decay chain of "green" series biexcitons [191]. Now, most of these features are interpreted as transitions from yellow excitons to the ground electronic state occupied by phonons [192]. A few lines still remain unexplained and may be caused by unidentified impurities [193].

Despite the lack of existing experimental evidence for biexcitons in Cu₂O, recombination of exciton pairs by Auger scattering has been invoked to explain exciton kinetics [194, 195]. In Auger scattering, one exciton in a pair is annihilated while the second exciton is ionized. Auger scattering is believed to be responsible for suppressing the Bose-Einstein condensation of excitons and for creating relaxation "explosions" in ultracold traps in Cu₂O [195, 196, 197]. While measured exciton density-dependent dynamics are consistent with the Auger model [198], the values of the Auger proportionality coefficient reported by different experimental sources vary by two orders of magnitude [199]. The variation is explained in part by differences in the total exciton number, the volume occupied inhomogeneously by the excitons, the exciton temperature [194], strain [200], and spin state [194, 201].

In general, there are two Auger mechanisms: one in which the exciton pair forms a bound

state before decaying (i.e. biexciton-Auger) and one in which it does not (i.e. free exciton-Auger) [194]. When the exciton temperature T_{Ex} is low compared to the biexciton binding energy ΔE_{Bx} , i.e. $k_B T_{Ex} \ll \Delta E_{Bx}$, the condition relevant for Bose-Einstein condensation [195, 202], we expect the biexciton-Auger behavior to dominate. When the exciton temperature is high compared to the biexciton binding energy, i.e. $k_B T_{Ex} \gg \Delta E_{Bx}$, the condition relevant for most ultrafast and high intensity experiments, we expect the free exciton-Auger behavior to dominate [203].

In the free exciton-Auger model, the decay rate of the exciton density n is expected to be linearly proportional to the exciton temperature [204], i.e. $dn/dt \propto -T_{Ex}n^2$, which is qualitatively consistent with the results at high T_{Ex} [205]. In the biexciton-Auger model, the decay rate is expected to be inversely proportional to the exciton temperature, i.e. $dn/dt \propto -n^2/T_{Ex}$, as has been previously observed at low T_{Ex} [206]. This indirect evidence is the strongest indication for the existence and significant influence of biexcitons in Cu₂O. Thus, direct observation of biexcitons and measurement of the biexciton binding energy is necessary to support any biexciton-based model of exciton kinetics and resolve differences among Auger kinetics measurements [200, 207, 208, 209, 210, 211, 212].

7.2 Cuprous Oxide Crystal Growth

A Cu₂O crystal was grown by our collaborators using their previously reported method for producing relatively large, phase pure crystals with minimal point defects [213, 214, 215]. A 0.9999 pure Cu rod with a 6.35 mm diameter was oxidized at 1045 °C in air for 3 days with a heating and cooling rate of 5 °C/min. It was crystallized in air at a rate of 3.5 mm/hr in an optical floating zone furnace. The feed and support rods were counter-rotating at 7 RPM. A 2 cm section of the crystal was annealed at 1045 °C for 5 days with a heating and cooling rate of 5 °C/min. The crystal was then cut perpendicular to the [110] axis, the relative orientation of the laser polarization in our measurements. A slice of the oriented



Figure 7.1: Two-dimensional rephasing spectrum of cuprous oxide at 10 K. (a) Rephasing spectrum at 10 K. (b) Representative Feynman diagrams for orthoexciton-polariton and phonon-assisted orthoexciton pathways, where red, orange, and yellow arrows denote field interactions involving the changes in phonon occupation number by -1, 0, and +1 which are nominally at photon energies of 2018, 2032, and 2046 meV, respectively. See Appendix B for full elucidation of Feynman diagrams. (c) Energy level diagram for orthoexcitons.

crystal was polished, resulting in a sample 0.401 mm thick. For the experiment, the Cu_2O crystal was cooled to 10 K in a Janis ST-100-H cryostat and controlled by a Lake Shore 331 Temperature Controller.

7.3 One-Quantum Rephasing Spectrum

To accurately identify the features of the two-quantum spectrum, we first discuss the onequantum (1Q) optical response of the 1s yellow orthoexciton revealed by the two-dimensional (2D) rephasing spectrum. The data gave no evidence of any contribution from the optically forbidden paraexciton. Figure 7.1a shows the rephasing spectrum with four qualitatively distinct features (locations A, C, G, and I).

The two diagonal features near 2046 meV (C) and 2018 meV (G) are approximately 14 meV from the 2032 meV orthoexciton-polariton energy, E_{Ex} . This energy difference corresponds to the Γ_{12}^- optical phonon energy, E_{ph} [192]. Exciton-phonon coupling has been shown to substantially aid optical excitation by phonon creation or annihilation [196]. Therefore,



Figure 7.2: Photoluminescence spectrum. Pumping at 3100 meV and detecting polarizations parallel to [110] reveals the weak, higher energy exciton-polariton emission and the strong, lower energy phonon-assisted emission (i.e. phonon creation).

the 2046 meV diagonal peak is consistent with a transition between the ground vibronic state $|g, 0\rangle$ and an orthoexciton state with one Γ_{12}^- phonon $|e, 1\rangle$ (i.e. phonon creation: $E_{Ex} - E_{ph}$). The 2018 meV diagonal peak is consistent with a transition between a phonon-populated ground electronic state $|g, 1\rangle$ and the ground vibrational state of the orthoexciton $|e, 0\rangle$ (i.e. phonon annihilation: $E_{Ex} - E_{ph}$). The 2046 meV absorption is much stronger because it originates from the ground vibronic state while the 2018 meV transition requires at least one phonon in the ground electronic state, which is less probable for experiments at 10 K but possible due to a large phonon density of states [196]. As expected, the 2032 meV orthoexciton-polariton diagonal peak (E) does not distinctly appear because it is a weakly allowed quadrupole resonance.

The two off-diagonal features with emission near 2018 meV (A) and 2046 meV (I) are crosspeaks between the phonon-assisted diagonal peaks (C and G). However, the 2018 meV crosspeak is shifted toward the 2032 meV orthoexciton-polariton emission energy (B). One reason for the shift is non-zero kinetic energy of the phonon-assisted orthoexcitons [196]. Due to the flatness of the phonon dispersion around the gamma point, phonon-assisted orthoexcitons can gain kinetic energy from high wavevector phonons, resulting in a high energy tail along both the absorption and emission axes. The higher energy tail is also observed in the photoluminescence spectrum in figure 7.2, where the strong, lower energy $E_{Ex} - E_{ph}$ emission is phonon-assisted and creates a phonon on the ground electronic state $(|e, 0\rangle \rightarrow |g, 1\rangle)$ and the weak, higher energy E_{Ex} emission is purely excitonic $(|e, 0\rangle \rightarrow |g, 0\rangle)$. A second possible reason is weak crosspeaks (B, D, F, and H) between the orthoexcitonpolariton and phonon-assisted orthoexcitons. These crosspeaks gain oscillator strength from the phonon-assisted transition dipole moments and may contribute to the overall spectrum even though the orthoexciton-polariton diagonal peak (E) is not directly observed.

7.4 Two-Quantum Spectrum

Based on the participation of the Γ_{12}^- phonon in the 1Q spectrum, we now discuss the twoquantum (2Q) spectrum shown in Figure 7.3a. First, we note there are strong 2Q coherences that unambiguously demonstrate significant two-exciton correlations. Additionally, the spectral density lies almost entirely below the y = 2x line, indicating the bi-orthoexciton is lower in energy than twice the orthoexciton energy and thus possesses a finite binding energy, $\Delta E_{Bx} = 2E_{Ex} - E_{Bx}$. However, a quantitative estimate of the bi-orthoexciton binding energy requires detailed analysis of the exciton-phonon coupling.

In the 2Q spectrum, we observe a major and a minor feature. The major feature emits near 2046 meV and is consistent with a phonon-assisted orthoexciton transition in which a phonon is annihilated upon emission $(E_{Ex} + E_{ph})$. The 2Q energy of the major feature is centered at 4066 meV and has a FWHM of approximately 80 meV but extends over more than 100 meV. Under the assumptions that at most one phonon can be created or annihilated per field interaction and the initial transition originates from the ground electronic state with 0 or 1 phonons (based on the 1Q spectrum), we can identify three excitation paths from the ground electronic state $|g, 0\rangle$ into a bi-orthoexciton coherence with 0, 1, and 2 phonons $|b, n\rangle$ that can emit with transition $E_{Ex} + E_{ph}$. Neglecting the biexciton binding energy, these three



Figure 7.3: Two-dimensional two-quantum spectrum of cuprous oxide at 10 K. (a) Twoquantum spectrum at 10 K. (b) Representative Feynman diagrams for orthoexciton-polariton and phonon-assisted orthoexciton pathways, where red, orange, and yellow arrows denote field interactions involving the changes in phonon occupation number by -1, 0, and +1 which are nominally at photon energies of 2018, 2032, and 2046 meV, respectively. See Appendix C for full elucidation of Feynman diagrams.

paths give rise to signals at $2E_{Ex}$, $2E_{Ex} + E_{ph}$, $2E_{Ex} + 2E_{ph}$, which are indicated in figure 7.3a at locations α , β , and γ , respectively. These three features have similar oscillator strength and overlap to form one broad peak centered at the average 2Q energy: $2E_{Ex} + E_{ph} - \Delta E_{Bx}$. The 2Q spectrum shows the corresponding spectral feature is shifted down in 2Q energy by some biexciton binding energy ΔE_{Bx} . The location of the major feature indicates a biorthoexciton binding energy of 12 meV, within the range of theoretically predicted values [190, 216, 217].

To better understand the determination of the biexciton binding energy, figure 7.4 illustrates the analysis for extracting ΔE_{Bx} from three overlapping 2Q signals with the same emission energy. Figure 7.4a illustrates the case where the non-overlapping signals of three unbound biexcitons at 2Q energy $2E_{Ex} + 2E_{ph}$, $2E_{Ex} + E_{ph}$, and $2E_{Ex}$ emit from $E_{Ex} + E_{ph}$. We observe there are three distinct peaks each separated in 2Q energy by the phonon energy E_{ph} . Figure 7.4b demonstrates how the three biexciton peaks shift to lower 2Q energy for a positive biexciton binding energy. Figure 7.4c illustrates the case where the overlapping signals of three unbound biexcitons at 2Q energy $2E_{Ex} + 2E_{ph}$, $2E_{Ex} + E_{ph}$, and $2E_{Ex}$ emit



Figure 7.4: Three overlapping two-quantum vibronic peaks. Bi-orthoexciton binding energy illustrations for the 3 overlapping vibronic signals of the 2Q phonon-assisted emission peak.

from $E_{Ex} + E_{ph}$. In this case, the three distinct peaks in figure 7.4a become one peak elongated along the 2Q energy direction. The center of the overlapping peak no longer lies on the y = 2x line even for the case of unbound biexcitons but is situated at the average 2Q energy of the three overlapping peaks: $E'_{Bx} = 2E_{Ex} + E_{ph}$. Figure 7.4d demonstrates how the elongated biexciton peak shifts to lower 2Q energy for a positive biexciton binding energy. The energy difference between the y = 2x line and the center of the elongated peak is now $\Delta E'_{Bx} = \Delta E_{Bx} + E_{ph}$. Thus, the energy difference between the y = 2x line and the elongated biexciton peak must be adjusted by E_{ph} to determine ΔE_{Bx} for three overlapping vibronic peaks.

The minor feature emits near 2032 meV and is consistent with orthoexciton-polariton emission (E_{Ex}) . The 2Q energy of the minor feature is centered at 4043 meV and has a FWHM of approximately 50 meV but extends over approximately 90 meV. We can identify three excitation paths that can emit with transition energy E_{Ex} : one from the phononpopulated ground electronic state $|g, 1\rangle$ into a bi-orthoexciton coherence including zero phonons $|b, 0\rangle$ and two from the ground vibronic state $|g, 0\rangle$ into a bi-orthoexciton coherence with 0 or 1 phonons $|b, n\rangle$. Neglecting the biexciton binding energy, these three paths give rise to signals with 2Q energies of $2E_{Ex} - E_{ph}$, $2E_{Ex}$, and $2E_{Ex} + E_{ph}$, which are indicated in figure 7.3a at locations δ , ε , and ζ , respectively.

However, there is no substantial signal intensity on or above the y = 2x line at the E_{Ex} emission energy and the signal is much narrower than the phonon-assisted emission peak. This suggests the bi-orthoexciton with one phonon (ζ) is not observed in the 2Q spectrum because of destructive interference with the excited state emission from pathway α . Thus, we postulate the minor feature originates mostly from the overlapping signals of the two bi-orthoexciton ground vibrational state pathways (δ and ε). These two features have similar oscillator strength and overlap to form one peak centered at the average energy: $2E_{Ex} - \frac{1}{2}E_{ph} - \Delta E_{Bx}$. The location of the minor feature indicates a bi-orthoexciton binding energy of 13 meV, which is in agreement with the binding energy of the phonon-assisted

peak.

To better understand the determination of the biexciton binding energy, figure 7.5 illustrates the analysis for extracting ΔE_{Bx} from two overlapping 2Q signals with the same emission energy. Figure 7.5a illustrates the case where the non-overlapping signals of two unbound biexcitons at 2Q energy $2E_{Ex}$ and $2E_{Ex} - E_{ph}$ emit from E_{Ex} . We observe there are two distinct peaks separated in 2Q energy by the phonon energy E_{ph} . Figure 7.5b demonstrates how the two biexciton peaks shift to lower 2Q energy for a positive biexciton binding energy. Figure 7.5c illustrates the case where the overlapping signals of two unbound biexcitons at 2Q energy $2E_{Ex}$ and $2E_{Ex} - E_{ph}$ emit from E_{Ex} . In this case, the two distinct peaks in figure 7.5a become one peak elongated along the 2Q energy direction. The center of the overlapping peak no longer lies on the y = 2x line even for the case of unbound biexcitons but is situated at the average 2Q energy of the two overlapping peaks: $E'_{Bx} = 2E_{Ex} - \frac{1}{2}E_{ph}$. Figure 7.5d demonstrates how the elongated biexciton peak shifts to lower 2Q energy for a positive biexciton binding energy. The energy difference between the y = 2x line and the center of the elongated peak is now $\Delta E'_{Bx} = \Delta E_{Bx} + \frac{1}{2}E_{ph}$. Thus, the energy difference between the y = 2x line and the elongated biexciton peak must be adjusted by $\frac{1}{2}E_{ph}$ to determine ΔE_{Bx} for two overlapping vibronic peaks.

From the lineshapes of the 2Q spectrum, we can also determine a bi-orthoexciton coherence lifetime of approximately 70 fs.

7.4.1 Feynman Diagram Analysis

To validate our assumptions about the number and relative strengths of the 2Q signals, we construct the set of all possible pathways, each presented by a distinct double-sided Feynman diagram, that constitute the 2Q signal at each emission energy, including phonon creation (2018 meV), exciton-polariton emission (2032 emV), and phonon annihilation (2048 meV). We enumerate all the possible 2Q Feynman diagrams under the assumptions that a field interaction can create or annihilate at most one phonon and the initial transition originates



Figure 7.5: Two overlapping two-quantum vibronic peaks. Bi-orthoexciton binding energy illustrations for the 2 overlapping vibronic signals of the 2Q orthoexciton-polariton emission peak.

(a)		Non-Overlapping Pathways								Overlapping Pathways				
		Emission Energy (meV)								Emission Energy (meV)				
		2018		2032		2046				2004	2018	2032	2046	
Two-Quantum Energy (meV)	4092					γ'	γ	() N	92			γ'	γ	
						-81	81	(me	4			-85	85	
						-81*	81*	ergy	78		ζ'	Z	β	
	4078			ζ'	ζ	β'	β	Ene	6		-19	0	19	
				-18	18	-18	18	tum	64	θ'	Θ	E	α	
				-18*	18*	-18*	18*	uan	40	-99	-2	2	99	
	4064	θ'	θ	ε΄	ε	α'	α	0-0'	50	η'	N	δ		
		-90	0	-10	10	0	90	μ	4	-1	0	1		
		-171*	171*	-19*	19*	-171*	171*							
	1050	η'	η	δ'	δ									
		0	0	0	0									
		-18*	18*	-18*	18*									

Figure 7.6: Graphical displays of two-quantum signal via Feynman diagram analysis. Graphical displays for non-overlapping (a) and overlapping (b) pathway strengths for all possible 2Q Feynman diagrams. * indicates pathways involving a phonon-populated ground electronic state. ' indicates excited state absorption pathways.

from the ground electronic state with 0 or 1 phonons. All the 2Q Feynman diagrams are enumerated in Appendix C.

For our qualitative estimate of the 2Q signal strengths, we assign the orthoexcitonpolariton transition dipole moment a strength of 1 and the phonon-assisted transition dipole moment a strength of 3. This 3:1 ratio is estimated from the photoluminescence spectrum in figure 7.2, which demonstrates a 9:1 ratio of the phonon-assisted to orthoexciton-polariton emission intensity. We also assume the transition dipole moments between the ground and exciton states and between the exciton and biexciton states are equivalent. For pathways originating from the phonon-populated ground electronic state, we assume 5% of electrons experience sufficient thermal excitation at 10 K, due to the large phonon density of states, to involve a phonon. The strength of a single pathway is given by the product of the four transition dipole moments and the initial thermal population. If multiple pathways exist for the same signal, they are summed to produce the total signal strength.

Each 2Q signal strength is graphically organized in figure 7.6a first assuming there is

no signal overlap. A plain Greek letter X indicates the stimulated emission pathway at a given 2Q and emission energy. X' indicates the excited state absorption pathway, which is shifted by the biexciton binding energy (13 meV) along the emission axis. A plain number Y indicates the total strength of transitions from the ground vibronic state, while Y* indicates the total strength of transitions involving an optical phonon on the ground electronic state.

Because the biexciton binding energy (13 meV) is close to the optical phonon energy (14 meV), the excited state absorption (ESA) of a given pathway should destructively interfere with the stimulated emission (SE) of the pathway at the next lowest emission energy. The destructive interference is due to the SE and ESA pathways being out of phase with each other. This can readily seen in the SE and ESA polarizations generated by the three excitation pulses:

$$P_{SE}(t') = \left(\frac{-i}{\hbar}\right)^{3} \int_{0}^{\infty} dt \int_{0}^{\infty} dT \int_{0}^{\infty} d\tau \overrightarrow{E}_{3}^{*}(t'-t) \overrightarrow{E}_{2}(t'-t-T) \overrightarrow{E}_{1}(t'-t-T) \times Tr\left(\left[\left[\overrightarrow{\mu}_{eg}(T+t), \overrightarrow{\mu}_{be}(T)\right], \overrightarrow{\mu}_{eb}(0)\right], \overrightarrow{\mu}_{ge}(0)\right] \rho(0)\right)$$
(7.1)

$$P_{ESA}(t') = \left(\frac{i}{\hbar}\right) \left(\frac{-i}{\hbar}\right)^2 \int_0^\infty dt \int_0^\infty dT \int_0^\infty d\tau \overrightarrow{E}_3^*(t'-t) \overrightarrow{E}_2(t'-t-T) \overrightarrow{E}_1(t'-t-T) \times Tr\left(\left[\left[\left[\overrightarrow{\mu}_{be}(T+t), \overrightarrow{\mu}_{ge}^*(T)\right], \overrightarrow{\mu}_{eb}(0)\right], \overrightarrow{\mu}_{ge}(0)\right] \rho(0)\right)$$
(7.2)

Figure 7.6b shows the expected 2Q signal strengths if neighboring stimulated emission and excited state absorption pathways perfectly cancel. We observe the excited state absorption pathway of peak β exactly cancels with the stimulated emission pathway of ζ (figure 7.6b, position Z).

$$E_{SE}^{\zeta} (4078 \ meV, \ 2032 \ meV) = E_{ESA}^{\beta} (4078 \ meV, \ 2046 \ meV - \Delta E_{Bx})$$
(7.3)

Thus we expect to see little or no spectral density above the y = 2x line at the 2032

meV emission energy. This is consistent with the measured 2Q spectrum and supports our assumption that only the overlapping E and δ pathways in figure 7.6b contribute to the minor feature. Additionally, the strength of the overlapping signal at E is 2, while the strength at δ is 1. These two signal strengths are similar, so the minor feature in the 2Q spectrum should be nearly symmetric about the two pathways.

The strength of the signals at α and γ are comparable at 99 and 85, respectively. The near equivalence of the expected spectral densities of the stimulated emission pathways of α and γ in figure 7.6b supports our assumption that the major feature in the 2Q spectrum is symmetric about pathway β .

A similar Feynman diagram analysis was performed for the rephasing spectrum and can be found in Appendix B.

7.5 One- and Two-Quantum Spectrum Simulations

To corroborate our interpretation of the 1Q and 2Q spectrum, the third-order response functions of Cu_2O were simulated using the density matrix formalism and assuming a 13 meV biexciton binding energy [41]. The third-order response functions of 2D spectra are discussed in more detail in chapter 4. Here, we merely give the expression for the matrix elements of the rephasing and 2Q response functions, including the ground state bleach (GSB), stimulated emission (SE), and excited state absorption (ESA) terms:

$$R_{R:GSB}^{(3)}\left(\omega_{ge}, T, \omega_{eg}\right) = \rho\left(0\right) \left|\mu_{ge}\right|^4 \exp\left[-\left(i\omega_{ge} + \Gamma_{ge}\right)\tau\right] \exp\left[-\left(i\omega_{eg} + \Gamma_{eg}\right)t\right]$$
(7.4)

$$R_{R:SE}^{(3)}\left(\omega_{ge}, T, \omega_{eg}\right) = \rho\left(0\right) \left|\mu_{ge}\right|^{4} \\ \times \exp\left[-\left(i\omega_{ge} + \Gamma_{ge}\right)\tau\right] \exp\left[-\Gamma_{ee}T\right] \exp\left[-\left(i\omega_{eg} + \Gamma_{eg}\right)t\right] \quad (7.5)$$

$$R_{R:ESA}^{(3)}\left(\omega_{ge}, T, \omega_{be}\right) = -\rho\left(0\right) \left|\mu_{ge}\right|^{2} \left|\mu_{eb}\right|^{2} \\ \times \exp\left[-\left(i\omega_{ge}+\Gamma_{ge}\right)\tau\right] \exp\left[-\Gamma_{ee}T\right] \exp\left[-\left(i\omega_{be}+\Gamma_{be}\right)t\right]$$
(7.6)

$$R_{2Q:SE}^{(3)}\left(\omega_{gb,}\omega_{ge}\right) = \rho\left(0\right)\left|\mu_{ge}\right|^{2}\left|\mu_{eb}\right|^{2}\exp\left[-\left(i\omega_{gb}+\Gamma_{gb}\right)T_{2Q}\right]\exp\left[-\left(i\omega_{ge}+\Gamma_{ge}\right)t\right]$$
(7.7)

$$R_{2Q:ESA}^{(3)}(\omega_{gb},\omega_{eb}) = -\rho(0) |\mu_{ge}|^2 |\mu_{eb}|^2 \exp\left[-\left(i\omega_{gb} + \Gamma_{gb}\right)T_{2Q}\right] \exp\left[-\left(i\omega_{eb} + \Gamma_{eb}\right)t\right]$$
(7.8)

where ω_{ij} is the resonance frequency between states *i* and *j*, μ_{ij} is the transition dipole moment, and Γ_{ij} is the dephasing time. An asymmetric Lorentzian lineshape was used to reproduce the high-energy tails of the phonon-assisted peaks observed in the photoluminescence in figure 7.2 [218]. In this model, the full width at half maximum (γ) of the Lorentzian lineshape function $f(\omega)$ varies as a function of frequency:

$$\gamma(\omega) = \frac{\gamma_0}{1 + \exp\left[a\left(\omega - \omega_0\right)\right]} \tag{7.9}$$

$$f(\omega) = \frac{1}{\pi \gamma(\omega)} \left[\frac{\gamma^2(\omega)}{(\omega - \omega_0)^2 + \gamma^2(\omega)} \right]$$
(7.10)

where γ_0 is the full width at half maximum at the center frequency ω_0 and a is a measure of the asymmetry of the lineshape. These simulations are qualitative in nature and only attempt to reproduce the general features of the rephasing and 2Q spectra.

The simulation of the rephasing spectrum is shown in figure 7.7a. The calculation of the response function reproduces the relative spectral intensities of the phonon creation (2046 meV) and phonon annihilation (2018 meV) absorption paths. The high-energy tails are also clearly visible. The [2046 meV, 2018 meV] crosspeak (figure 7.1a, peak A) is clearly shifted toward the 2032 meV orthoexciton-polariton emission energy, while the [2018 meV, 2046 meV] crosspeak (C) is shifted slightly away. The off-diagonal peak associated with phonon annihilation at [2018 meV, 2046 meV] (I) is weaker in the simulation then in the experimental data, but the diagonal peak at [2018 meV, 2018 meV] (G) is reproduced fairly well. Overall, the simulation qualitatively reproduces the data well enough to support our interpretations



Figure 7.7: Rephasing and two-quantum spectra simulations. Simulations of the third-order response function for the rephasing (a) and 2Q spectra (b) using a combination of symmetric and asymmetric Lorentzian lineshapes and assuming the experimentally determined 13 meV bi-orthoexciton binding energy.

of the 1Q spectrum and to validate our assumptions for the analysis of the 2Q spectrum.

In figure 7.7b, we show the simulation of the 2Q spectrum. The calculation reproduces the 2046 meV 2Q signal emission fairly well. However for the 2032 meV 2Q signal emission, there is some simulated spectral intensity above the y = 2x line, possibly due to incomplete destructive interference of the 4078 meV peak with the excited state absorption pathway from the [4078 meV, 2046 meV] peak (figure 7.3a, peak α). The 2Q energy of the 2032 meV emission is also only slightly lower in 2Q energy than the 2046 meV emission, possibly due to weak signal from the phonon-populated ground electronic state pathway (δ). The 2032 meV emission is also more intense compared to the 2046 meV emission. The lower emission energy features are mainly from the excited state absorption pathways of the 2032 meV 2Q signal emission, which we do not observe experimentally. Overall, the simulation qualitatively reproduces the major feature of the 2Q spectrum but breaks down for the minor feature. We are confident in our interpretation of the experimental data, particularly of the major feature, but quantitative simulation of the 2Q spectrum requires many unknown and difficult to optimize parameters, including some dephasing rates and the difference between



Figure 7.8: Temperature-dependent, integrated Maxwell-Boltzmann distribution of biexciton energies. Temperature-dependent population fraction of bi-orthoexcitons with thermal energy less than the 13 meV bi-orthoexciton binding energy.

the ground to exciton and exciton to biexciton transition dipole moments.

7.6 Temperature-Dependent Experiments

Temperature-dependent 2Q spectra were recorded in the 10-200 K range to further study the bi-orthoexcitons in Cu_2O . Figure 7.10 shows the normalized integrated signal intensity of the entire 2Q spectrum as a function of temperature. As the temperature of the system increases and the kinetic energy of the bi-orthoexcitons exceeds their binding energy, the total 2Q signal intensity decreases. A Maxwell-Boltzmann distribution can be used to model the population of bi-orthoexcitons below the binding energy as a function of temperature:

$$f_{E<\Delta E_{Bx}}^{MB}(T) = \int_{0}^{\Delta E_{Bx}} f^{MB}(E,T) dE$$
(7.11)

$$f^{MB}(E,T) = 2\left(\frac{1}{k_BT}\right)^{\frac{3}{2}}\sqrt{\frac{E}{\pi}}\exp\left[\frac{-E}{k_BT}\right]$$
(7.12)

Figure 7.8 shows the calculation of the percentage of the bi-orthoexciton thermal popu-



Figure 7.9: Laser intensity at the temperature-dependent exciton energy. (a) Temperaturedependent orthoexciton-polariton energy. (b) Laser spectrum used in the 2D experiments. (c) Temperature-dependent laser intensity squared at 2046 meV, $I_{Laser}^2(E_{Ex}(T) + E_{ph})$.

lation below the bi-orthoexciton binding energy (13 meV) as a function of temperature using equation 7.11. It is assumed any biexciton with a thermal energy equal to or greater than the biexciton binding energy $(E_{Bx}(T) \ge \Delta E_{Bx})$ dissociates and does not contribute to the 2Q signal intensity.

The orthoexciton-polariton energy is also temperature-dependent and decreases with temperature. At high temperatures, the transition frequency begins to move out of the laser pulse bandwidth. A model developed by Itoh and Narita for the temperature dependence of the band gap energy $(E_{BG}(T))$ and the orthoexciton binding energy $(\Delta E_{Ex}(T))$ describes the orthoexciton-polariton energy as a function of the temperature [219]:

$$E_{Ex}(T) = E_{BG}(T) - \Delta E_{Ex}(T) + E_{O-P}$$
(7.13)

$$E_{BG}(T) = 2,195 \, meV - 26 \, meV \times \coth\left[\frac{E_{ph}}{2k_BT}\right]$$
(7.14)

$$\Delta E_{Ex}(T) = 150 \, meV + 0.64 \, meV \times \coth\left[\frac{E_{ph}}{2k_BT}\right] \tag{7.15}$$

where E_{O-P} is the ortho-para exchange splitting energy. We assume E_{O-P} is temperature independent because it is an order of magnitude smaller than $\Delta E_{Ex}(T)$ and two orders of magnitude smaller than $E_{BG}(T)$, both of which vary around 1% in this temperature range [219]. We can account for the temperature-dependent orthoexciton-polariton energy



Figure 7.10: Temperature-dependent bi-orthoexciton signal intensity. Model predictions are based on bi-orthoexciton binding energies of 11 (green), 13 (blue), and 15 (red) meV.

by normalizing the bi-orthoexciton signal intensity to the laser intensity squared at the energy of the strongest transition: $I_{Laser}^2 (E_{Ex} (T) + E_{ph})$. We assume E_{ph} is temperature independent because it is also two orders of magnitude smaller than $E_{Ex} (T)$ and is likely to only vary around 1%, as in other semiconductors in this temperature range [220, 221]. Figure 7.9a plots the temperature-dependent orthoexciton-polariton energy $E_{Ex} (T)$ using the Itoh-Narita model expressed in equation 7.13. Figure 7.9b shows the laser pulse spectrum used in the 2D experiments. Figure 7.9c shows the laser intensity squared at the strongest transition energy as a function of temperature, $I_{Laser}^2 (E_{Ex} (T) + E_{ph})$.

The effects of the bi-orthoexcitons thermal energy (figure 7.8) and the shifting of the orthoexciton energy out of the laser pulse bandwidth (figure 7.9c) are combined to model the temperature-dependent bi-orthoexciton 2Q signal intensity. We multiply the results from figures 7.8 and 7.9c to produce the model presented in figure 7.10. Figure 7.10 superimposes the temperature-dependent bi-orthoexciton signal intensity with the theoretical prediction of our model. For the 13 meV bi-orthoexciton binding energy determined in the 2Q spectrum at 10 K, the model agrees well with the temperature-dependent data. The deviation of the theoretical bi-orthoexciton signal intensity from the experimental intensity above 130 K can

be seen to arise from the non-Gaussian profile of the laser spectrum in figures 7.9a-c. The 2Q signal intensity must deviate from the expected behavior as the resonances move out of the laser pulse bandwidth and depend less on the laser pulse profile.

7.7 Conclusions

From the 2Q spectrum of Cu_2O , we directly observe two-exciton correlations. We measure a bi-orthoexciton binding energy of 12-13 meV and a coherence lifetime of approximately 70 fs. Our model of the temperature-dependent bi-orthoexciton signal intensity is also consistent with a 13 \pm 2 meV bi-orthoexciton binding energy. Our estimate is much greater than the 4.7 meV value calculated from third-order central cell corrected predictions [190] and the 6.2 meV value calculated with a variational method [216] but is in agreement with the 13 meV Feynman path integral prediction [217]. Overall, our results support the suggestion that biexcitons play a key role in Auger recombination of excitons below 150 K and thus in the suppression of exciton-polariton Bose-Einstein condensation.

Chapter 8

Exciton-Phonon and Exciton-Exciton Interactions in Monolayer Transition Metal Dichalcogenides

In chapter 8, we discuss preliminary experiments exploring the extent of exciton-exciton and exciton-phonon interactions in monolayer transition metal dichalcogenides. Our samples were grown using chemical vapor deposition by our collaborators at the National Tsing Hua University in the group of Yi-Hsien Lee. Characterization of the samples by microscopy and reflectance were performed by Edbert Sie at MIT in the group of Nuh Gedik. Twodimensional spectroscopy was performed by myself, Jake Siegel, and Yoseob Yoon.

8.1 Monolayer Transition Metal Dichalcogenides Physics

Recent advancements in the isolation of monolayer transition metal dichalcogenides (TMDs: MX_2) by exfoliation and chemical vapor deposition (CVD) have led to the discovery of new and exciting semiconductor physics [222, 223, 82, 85]. This class of materials is most commonly composed of the group VI transition metals (M) molybdenum (Mo) or tungsten (W) and the group XVI chalcogens (X) sulfur (S), selenium (Se), or tellurium (Te). Bulk TMDs

are characterized by strongly bound layers weakly coupled through van der Waals forces, similar to graphite. Each layer is three atoms thick and comprised of a layer of transition metals sandwiched between two layers of chalcogen atoms (i.e. XMX), illustrated in figure 8.1a. Each monolayer is arranged in a hexagonal structure that results in a hexagonal Brillouin zone with six K-point valleys, similar to the band structure of graphene. Unlike graphene, the chemical composition of TMDs leads to a finite band gap of approximately 1-2 eV. In the bulk, TMDs are indirect band gap semiconductors. In the monolayer limit, TMDs become direct band gap semiconductors, due to quantum confinement of their *d*-electrons, and exhibit strong absorption and photoluminescence [224, 83]. Monolayer TMDs also do not possess an inversion center, unlike bulk and even numbered multi-layer TMDs, and results in non-equivalent K and -K valleys in the Brillouin zone.

8.1.1 Emergence of Valley Polarization

The emergence of non-equivalent K valleys in monolayer TMDs has garnered significant interest because of the potential to carry quantum information in the valley indices of electrons and holes. In this design, electrons or holes of the same energy but binary positions in momentum space can be used to build "valleytronic" devices, similar to electronic and spintronic devices. The large atomic weight of the transition metal in TMDs also produces strong spin-orbit coupling that splits the valence and conduction bands into two bands each. The excitons formed by holes in each of the valence bands are clearly observed in the linear absorption spectrum (figure 8.1b) [225, 226]. The lower- and higher-energy excitons are commonly referred to as A and B, respectively.

The spin-orbit coupling also couples the spin and valley states and allows a single spin or valley to be accessed via circularly polarized light due to stringent optical selection rules, summarized in figure 8.1c [84]. In the K valley, electrons from the higher (lower) energy valence band with positive $|\uparrow\rangle$ (negative $|\downarrow\rangle$) spin are selectively excited by right-handed σ^+ (left-handed σ^-) circular polarized light into the conduction band; while in the -K valley,



Figure 8.1: Monolayer transition metal dichalcogenides. (a) Hexagonal lattice structure, with one layer of transition metals (M: red) sandwiched between two layer of chalcogen atoms (X: blue). (b) Linear absorption spectrum of monolayer MoS₂, where the A and B exciton states absorb at 1.90 eV and 2.06 eV, respectively [98]. (d) Optical selection rules. Positive $|\uparrow\rangle$ and negative $|\downarrow\rangle$ spin electrons in the K and -K valleys are selectively excited from the upper (red) and lower (blue) valences bands into the conduction bands (green) by either right-handed σ^+ or left-handed σ^- circularly polarized light.



Figure 8.2: Coulombic interactions in two-dimensional materials. (a) In-plane Coulombic interactions are determined by the in-plane bulk dielectric constant ε_{TMD} , which is large and decreases electron (e^-) and hole (h^+) interactions as indicated by the curved electric field lines. (b) Out-of-plane Coulombic interactions are determined by the dielectric constants of the substrate ε_{sub} and vacuum ε_0 and the thickness of the TMD monolayer d.

electrons from the higher (lower) energy valence band with negative $|\downarrow\rangle$ (positive $|\uparrow\rangle$) spin are selectively excited by left-handed σ^- (right-handed σ^+) circular polarized light into the conduction band. Numerous experimental measurements have demonstrated the generation of robust valley polarizations of over 50% by circular dichroism lasting longer than 1 ns [227, 228, 229].

8.1.2 Enhancement of Coulombic Interactions

Monolayer TMDs are also of significant interest because of their unprecedented Coulombic interactions for inorganic semiconductors. Coulombic interactions in the plane of a monolayer are analogous to the in-plane interaction of the bulk due to their large dielectric constants. However, charged particles can also interact out of the plane and the interaction strength is dependent on the dielectric constants of the substrate and vacuum. This out-of-plane Coulombic interaction allows for much stronger charged particle correlations when the surrounding dielectric constants are small and screening of electrons and holes decreases, as depicted in figure 8.2. The dielectric constant of our sapphire substrates $\varepsilon_{sub} = 3.2$ reduces the interaction of charged particles through the substrate, but the small vacuum dielectric

Materials	ε	$\Delta E_X \text{ (meV)}$	ΔE_{X^-} (meV)	$\Delta E_{XX} \ (meV)$
$MoS_2 WL$	11.6 [86]	850 [230]	20 [98]	70 [231]
$WS_2 ML$	13.0 [86]	710 [232]	30 [233]	65 [234]
GaAs QW	12.3 [235]	9 [236]	1 [237]	1 [61]

Dielectric Constants and Multi-Particle Interactions

Table 8.1: Dielectric constants and multi-particle interactions in monolayer transition metal dichalcogenides. Monolayer (ML) MoS₂ and WS₂ and GaAs quantum well (QW) dielectric constants ε , exciton ΔE_X , trion ΔE_{X^-} , and biexciton ΔE_{XX} binding energies.

constant $\varepsilon_0 = 1$ allows strong charged particle interactions above the plane of the monolayer. The dielectric constants of monolayer MoS₂ and WS₂ and GaAs quantum wells are summarized in table 8.1 along with their exciton ΔE_X , trion ΔE_{X^-} , and biexciton ΔE_{XX} binding energies. The binding energies of all the multi-particle correlations in monolayer TMDs are more than an order of magnitude larger than those in quasi-two-dimensional quantum wells, demonstrating significantly stronger Coulombic interactions in the atomic limit.

The Hamiltonian of charged particles in a two-dimensional material on a substrate can easily written down.

$$\hat{H} = -\frac{1}{2\mu} \nabla_r^2 - \hat{V}_{2D} \left(\overrightarrow{r} \right)$$
(8.1)

where $\mu^{-1} = m_e^{-1} + m_h^{-1}$ is the reduced mass of the exciton and m_e and m_h are the masses of an electron and hole, respectively. The Coulomb potential \hat{V}_{2D} is characterized by the dielectric constants of the monolayer ε_{TMD} , substrate ε_{sub} , and vacuum ε_0 and takes the form:

$$\hat{V}_{2D}\left(\overrightarrow{r}\right) = \frac{\pi e^2}{\left(\varepsilon_{sub} + \varepsilon_0\right) r_0} \left[\frac{\overrightarrow{r}}{r_0} \left(H_0 - Y_0\right)\right]$$
(8.2)

$$r_0 = \frac{\varepsilon_{TMD}d}{\varepsilon_{sub} + \varepsilon_0} \tag{8.3}$$

where e is the charge of an electron, r_0 is the screening length that is dependent on d the thickness of the TMD monolayer (6.5 Å in MoS₂), and H_0 and Y_0 are the Struve function and the Bessel function of the second kind, respectively [238]. At long range, the Coulombic interaction behaves like a screened potential 1/r; while at short range, the Coulombic interaction has a weaker logarithmic divergence and allows for stronger charged-particle correlations. The transition between short and long range interactions is determined by the screening length r_0 .

For photovoltaic devices, strong electron-hole interactions are unfavorable because excitons must be dissociated, usually at an interface, in order to extract charge and generate a current. Thus, the large binding energies of excitons in monolayer TMDs make them less suited for light-harvesting devices. However, the high charge-carrier mobilities in monolayer TMDs are highly favorable and allow for fabrication of larger devices, the size of which can increase their quantum efficiency [239].

8.2 Monolayer Transition Metal Dichalcogenide Characterization

Prior to performing two-dimensional (2D) spectroscopy experiments to investigate excitonphonon and exciton-exciton interactions in monolayer TMDs, we characterized our samples using microscopy and reflection measurements. First, we took microscope images of our CVD-grown monolayer WS₂ sample on sapphire, shown in figure 8.3a. From the image, we can clearly distinguish the characteristic triangular single domains (light green) approximately 20 μ m across with a nucleation site (white) in the center. In this sample, many domains have grown together, forming a monolayer film covering the majority of the the substrate (dark green). Second, we used a microscope to measure the reflectance of white light off single domains of monolayer WS₂ at 10 K. Using the Kramers-Kronig relations, we calculated the linear absorption spectrum from the reflectance [240, 241]. Figure 8.3b shows a representative linear absorption spectrum of a single domain. In this spectrum, the absorption is centered at 2.07 eV with a full width at half maximum (FWHM) of approximately 0.04 eV. However, the central energy of the linear absorption changed significantly from domain to domain with greater than 0.02 eV variation, more than half the FWHM. The



Figure 8.3: Monolayer transition metal dichalcogenide characterization. (a) Microscopic image of CVD-grown monolayer WS_2 on sapphire with 100x magnification at 295 K. Light green regions indicate monolayer WS_2 , dark green regions the substrate, and white regions nucleation sites for single domains. (b) Representative linear absorption spectrum of a single domain of monolayer WS_2 at 10 K. The central energy of the absorption changes more than 0.02 eV in different domains, while the 0.04 eV full width at half maximum and lineshape remains relatively constant.

FHWM was relatively constant from domain to domain. The lineshape was also relatively constant, but some domains exhibited small shoulders that were redshifted 0.04 eV from the main peak, which could be due to trion or biexciton formation [233, 234].

8.3 Exciton-Phonon Interactions in Monolayer Transition Metal Dichalcogenides

The interaction of excitons and phonons is strongly dependent on the structure and dimensionality of the material. As the width of a material decreases and approaches the 2D limit, exciton-phonon coupling is predicted to significantly increase [242, 243]. Experiments using Raman spectroscopy to study the phonon modes in graphene, the most widely studied 2D material, have shown strong and layer-dependent electron-phonon interactions at and near the atomic limit [244, 245]. Thus, investigation of exciton-phonon coupling in mono-



Figure 8.4: Two-dimensional rephasing spectrum of monolayer MoS_2 at 295 K. (a) 2D rephasing spectrum of monolayer MoS_2 at 295 K. (b) Diagonal and anti-diagonal line cuts of the 2D rephasing spectrum reveal inhomogeneous and homogeneous linewidths of 130 and 10 meV, respectively.

layer TMDs is important to understand the interactions and dynamics of their elementary excitations.

8.3.1 Large Inhomogeneous Broadening in Chemical Vapor Deposition-Grown Monolayer Transition Metal Dichalcogenides

To study the interactions of excitons and phonons, we first used 2D rephasing spectroscopy to measure the homogeneous and inhomogeneous linewidths of the B exciton in monolayer MoS_2 at 295 K. Figure 8.4a clearly shows substantial inhomogeneous broadening in the rephasing spectrum. The homogeneous (i.e. anti-diagonal) and inhomogeneous (i.e. diagonal) lineshapes, plotted in figure 8.4b, are 10 and 130 meV, respectively. Substantial inhomogeneous broadening demonstrates that static disorder dominates the absorption linewidths of the monolayer. Static disorder generally occurs due to structural defects within a single domain (e.g. vacancies and interstitials) and variations from domain to domain (i.e. multiple domain illumination in the 30 μ m focus spot size) [246]. Dynamic disorder generally occurs due to exciton-phonon and exciton-exciton interactions and population recombination, which determines the homogeneous linewidth. Even at 295 K and large phonon population, the static rather than dynamic disorder is the major contributor to exciton dephasing. 2D rephasing measurements by Moody and co-workers also demonstrate significant inhomogeneous broadening in single domain monolayer WSe₂ at cryogenic temperatures [57]. Temperature-dependent experiments revealed exciton-phonon interactions due to acoustic phonon scattering were enhanced by a factor of 5 to 10 compared to quasi-2D GaAs quantum wells.

CVD-grown TMD monolayers generally exhibit more static disorder than exfoliated monolayers. Thus, exfoliated TMDs may reveal substantially less inhomogeneous broadening and significantly different 2D rephasing spectra. Large densities of structural defects can significantly hinder the electrical properties, such as charge transport. However, structural defects also increase the exciton radiative lifetime in monolayer TMDs due to exciton localization, which is beneficial for device fabrication [247].

8.3.2 Exciton Coupling to In-Plane Optical and Acoustic Phonon Modes

Second, we used 2D correlation spectroscopy to measure the primary phonons contributing to the dephasing of A excitons in monolayer WS₂ at 10 K. Figure 8.5a shows clear structure in the correlation spectrum along both the absorption and emission axes. Fitting the excitation energy-integrated projection of the correlation spectrum onto the emission axis reveals a progression of peaks equally separated by 22 meV with a FWHM of 26 meV, shown in figure 8.5b. Thripuranthaka M and Late conducted temperature-dependent Raman experiments and assigned their 22 meV peak to an optical/acoustic phonon combination band $E_{2g}^1 - LA(M)$, where E_{2g}^1 is an in-plane optical phonon mode at the Γ -point and LA(M) is an in-plane longitudinal acoustic phonon mode at the M-point [220]. The phonon progression observed in the correlation spectrum is not observed in the linear absorption spectrum. So it is unclear why the phonon progression is observed in the 2D spectrum, but it is clear



Figure 8.5: Two-dimensional correlation spectrum of monolayer WS₂ at 10 K. (a) Correlation spectrum of monolayer WS₂ at 10 K. (b) Excitation energy-integrated projection of the correlation spectrum amplitude onto the emission axis reveals a phonon progression of the 22 meV combination band $E_{2g}^1 - LA(M)$.

the $E_{2g}^1 - LA(M)$ combination phonon band is playing an important role in the optical response of excitons in monolayer WS₂ at 10 K. Temperature-dependent and population time-dependent measurements can further interrogate the role of phonons in modulating the excitonic properties of monolayer TMDs.

8.4 Exciton-Exciton Interactions in Monolayer Transition Metal Dichalcogenides

8.4.1 Biexciton Binding Energy via One-Quantum Measurements

From the absorptive (i.e. imaginary) component of the 2D correlation spectrum of A excitons in monolayer WS₂ at 10 K, we can determine the biexciton binding energy by the separation of the stimulated emission and excited state absorption peaks. Figure 8.6a illustrates the stimulated emission and excited state absorption Feynman diagrams. The stimulated emission pathway radiates from an exciton-ground state coherence $|e\rangle \langle g|$ with the exciton



Figure 8.6: Two-dimensional absorptive spectrum of monolayer WS₂ at 10 K. (a) Feynman diagrams for stimulated emission and excited state absorption pathways emitting at E_{Ex} and $E_{Ex} - \Delta E_{Bx}$, respectively. (b) 2D absorptive spectrum of monolayer WS₂ at 10 K with stimulated emission (positive) and excited state absorption (negative) peaks. (c) Excitation energy-integrated projection of the 2D absorptive spectrum amplitude onto the emission axis reveals a 45 ± 3 meV biexciton binding energy by the difference of the stimulated emission and excited state absorption peak emission energies.

energy E_{Ex} , while the excited state absorption pathway radiates from a biexciton-exciton coherence $|b\rangle \langle e|$ with energy $E_{Bx} - E_{Ex} = E_{Ex} - \Delta E_{Bx}$, where ΔE_{Bx} is the biexciton binding energy. Thus, the difference in emission energy between the stimulated emission and excited state absorption peaks is a measure of the biexciton binding energy. Figure 8.6b shows the 2D absorptive spectrum with the stimulated emission (negative) and excited state absorption (positive) peaks. The excitation energy-integrated projection of the 2D absorptive spectrum onto the emission axis, plotted in figure 8.6c, reveals the characteristic absorptive lineshape and a biexciton binding energy of 45 ± 3 meV, equivalent to a thermal energy of $\Delta E_{Bx}/k_B = 520K$.

8.4.2 Biexciton Dephasing Time via Two-Quantum Measurements

We also used two-quantum (2Q) spectroscopy to directly measure biexciton coherences and determine the AA biexciton binding energy and dephasing time in monolayer WS₂ at 10 K. The deviation of the 2Q signal from the y = 2x line directly reveals the biexciton binding energy ($\Delta E_{Bx} = 2E_{Ex} - E_{Bx}$) without interference from single-exciton signals. Figure 8.7 shows the 2Q spectrum and reveals a biexciton binding energy of 38 ± 8 meV, equivalent to a thermal energy of $\Delta E_{Bx}/k_B = 460K$. The linewidth of the 2Q signal also reveals a biexciton dephasing time faster than 100 fs. The low 2Q energy peak is a small contribution from the single-exciton signal in the 2Q spectrum.

The biexciton binding energies measured in the one-quantum absorptive and 2Q spectra are in good agreement and are more than an order of magnitude greater than the biexciton binding energies in GaAs quantum wells [61]. Power-dependent experiments by Moody and co-workers also revealed that exciton-exciton coupling in monolayer WSe₂ is enhanced by an order of magnitude or more compared to GaAs quantum wells [57]. Our measurements of unprecedentedly large biexciton binding energies are also in good agreement with other one-quantum measurements in monolayer TMDs [231, 234, 248]. Large biexciton binding energies make them relevant at room temperature and may allow them to significantly affect



Figure 8.7: Two-dimensional two-quantum spectrum of monolayer WS_2 at 10 K. (a) Twoquantum spectrum of monolayer WS_2 at 10 K. (b) Emission energy-integrated projection of the two-quantum spectrum amplitude onto the two-quantum axis reveals a biexciton dephasing rate faster than 100 fs.

the quantum efficiency of monolayer TMD-based devices.

8.5 Conclusions

From our preliminary experiments on monolayer TMDs, we were able to measure large inhomogeneous broadening in a CVD-grown monolayer TMD at 295 K, revealing that static disorder dominates the linear absorption spectrum and is the main contributor to exciton dephasing even at room temperature. 2D correlation spectroscopy also demonstrated strong contributions from the in-plane E_{2g}^1 optical and LA(M) acoustic phonon modes leading to modulation of the 2D spectrum. The absorptive component of the correlation spectrum and the two-quantum spectrum revealed an AA biexciton binding energy in monolayer WS₂ of 40-45 meV, orders of magnitude greater than quasi-two-dimensional GaAs quantum wells, and a biexciton dephasing time faster than 100 fs.

There are many more experiments planned for monolayer TMDs. First, we plan to measure the homogeneous and inhomogeneous linewidths of exfoliated TMD monolayers and investigate the contribution of static disorder to the linear absorption spectrum compared to CVD-grown monolayers. Second, we plan to use 2D correlation spectroscopy to elucidate the mechanisms of the exciton-phonon interaction by measuring temperature-dependent and population time-dependent 2D spectra. Third, we plan to use higher-order spectroscopies to measure 3-, 4-, ..., n-exciton interactions and determine the full extent of the multi-particle interactions, their binding energies, and coherence times. Finally, bilayers of different TMDs can also be synthesized to form heterostructures. We plan to investigate the extent of exciton-exciton interactions in these heterostructures and determine if interlayer biexcitons or higher-order correlations exist.
Chapter 9

Conclusions and Outlook

In this thesis, we discussed ultrafast spectroscopic experiments exploring the exciton dynamics and exciton-exciton and exciton-phonon interactions in organic and inorganic, nanostructured semiconductors. In chapter 5, we measured the weak coupling of excitons on the inner and outer walls of isolated and bundled double-walled J-aggregate nanotubes in solution. We determined exciton dephasing in solution-based nanotubes at room temperature is dominated by dynamic disorder, while glass nanotubes at cryogenic temperature is dominated by static disorder. We also observed photo-induced degradation and recovery in glass nanotubes at cryogenic temperature and the lowest possible excitation fluence, the mechanism for which is still being studied by our collaborators in the Bawendi group at MIT. In chapter 6, we utilized quantum process tomography to unambiguously characterize the secular processes of population decay and transfer and coherence decay, unequivocally determine the weak coupling of inner and outer wall excitons, and demonstrate the absence of non-secular processes in isolated J-aggregate nanotubes. In chapter 7, we made the first direct experimental measurement of biexcitons in cuprous oxide and provided evidence to support a biexciton-Auger mechanism that effectively suppresses Bose-Einstein condensation of excitons. In chapter 8, we discussed preliminary experiments exploring the strength of exciton-phonon and exciton-exciton interactions in monolayer transition metal dichalcogenides.

We plan to continue experiments exploring the exciton dynamics in monolayer transition metal dichalcogenides. First, we plan to measure the homogeneous and inhomogeneous linewidths of exfoliated monolayers and investigate the contribution of static disorder to the linear absorption spectrum as compared to chemical vapor deposition-grown monolayers. Second, we plan to elucidate the mechanisms of exciton-phonon interaction by measuring temperature-dependent and population time-dependent two-dimensional correlation spectra. Third, we plan to determine the full extent of multi-exciton correlations, their binding energies, and their coherence times using higher-order, multi-quantum spectroscopy. Finally, bilayers of different transition metal dichalcogenides can be synthesized to form unique heterostructures. We plan to investigate the extent of exciton-exciton interactions in these heterostructures and determine if interlayer multi-exciton correlations exist.

Another class of exciting nanoscale semiconductors is organic-inorganic hybrid materials such as quantum dot-J-aggregate systems [249, 250, 251]. Using sensitizing media to enhance optical and electronic properties is a common technique for improving the efficiency of devices. The improvement in efficiency is generally determined by the electronic coupling between the media. We plan to use two-dimensional spectroscopy to explore the coupling of excitons in quantum dot-J-aggregate hybrid systems.

The emergence of relatively high efficiency perovskite solar cells, such as methyl ammonium lead halides, have renewed interest in these semiconducting materials [252, 253]. The synthesis of new lead halide perovskite nanoplatelets has also garnered attention due to their tunability via quantum confinement [254, 255]. We plan to use multi-quantum spectroscopies to investigate the extent of multi-exciton correlations in this new class of nanoscale materials.

Exciton transport is also an important process in determining the efficiency of optical and electronic devices. More specifically, excitons in a photovoltaic device must diffuse to an interface in order to dissociate into electrons and holes and generate a current. If excitons cannot diffuse to an interface within their lifetime, charge separation does not efficiently occur and a substantial current cannot be generated. Excitons generally do not diffuse more than a micron within their lifetime, thus exciton transport should be measured directly on the nanoscale. However, most exciton transport measurements are made indirectly on the micro-scale [256]. Currently, we are working on performing transient grating measurements of exciton transport on the nanoscale. However, these experiments are hindered by excitonexciton annihilation even at the lowest possible excitation fluences. Thus, we are developing a new fluorescence-detected transient grating technique to further reduce the necessary excitation fluences while maintaing high spatial resolution.

The ultrafast spectroscopic experiments performed and proposed in this thesis are designed to investigate the interactions and time evolution of elementary excitations in organic and inorganic nanostructured semiconductors. Elucidating the ultrafast and nanoscale dynamics of excitons and phonons is crucial to understanding the physics of optical and electronic devices operating on these length and timescales. The results in this thesis will hopefully help guide the design of next generation optical and electronic devices utilizing nanoscale semiconductors.

Appendix A

Transient Grating Feynman Diagrams for Quantum Process Tomography

Appendix A enumerates all the possible transient grating Feynman diagrams for quantum process tomography of the inner and outer wall excitons of double-walled J-aggregate nanotubes. The Feynman diagrams are organized by experiment XYZ ($X, Y, Z \in I, O$), where the initial state $|X\rangle \langle Y|$ is selectively excited and the state $|Z\rangle$ is interrogated. During the delay T, we allow secular and non-secular processes to occur or any initial state $|X\rangle \langle Y|$ to evolve into any other final state $|X'\rangle \langle Y'|$ and determine the emission energy associated with the corresponding process matrix elements $\chi_{X'Y'XY}$ for each experiment.







Experiment: III χ_{IIII} χ_{OIII} χ_{oon} $\chi_{\scriptscriptstyle IOII}$ Simulated Emission $|g\rangle\langle g|$ $|g\rangle\langle g|$ $|g\rangle\langle I$ $|g\rangle\langle O$ $|I\rangle\langle I|$ $|I\rangle\langle O$ Т Т $|I\rangle\langle I|$ $|\overline{I}\rangle\langle\overline{I}$ $|g\rangle\langle g|$ $|g\rangle\langle g|$ E_I E_o **Excited State Absorption** χ_{oon} χ_{IIII} χ_{OIII} $\chi_{\scriptscriptstyle IOII}$ $|O\rangle\langle O|$ $|O\rangle\langle O|$ $|I\rangle\langle I|$ $|I\rangle\langle I|$ $I\rangle\langle II$ $|O\rangle\langle OI$ $|0\rangle\langle II$ $I\rangle\langle OI$ $|0\rangle\langle 0|$ $|I\rangle\langle I$ $|O\rangle\langle I|$ $|I\rangle\langle O|$ Т Τ Т Т $|I\rangle\langle I$ $|I\rangle\langle I|$ $|I\rangle\langle I|$ $|I\rangle\langle I$ 7 $|g\rangle\langle g|$ 1 $|g\rangle\langle g|$ 1 $|g\rangle\langle g|$ $|g\rangle\langle g|$ $E_I - \Delta E_{IO}$ $E_I - \Delta E_{II}$ $(2E_I - \Delta E_{II})$ $-E_o$ $E_o - \Delta E_{IO}$



 $\chi_{{}_{IIOI}}$ χ_{OIOI} χ_{00001} χ_{1001} Simulated Emission $|g\rangle\langle g|$ $|g\rangle\langle g|$ $|g\rangle\langle I$ $|g\rangle\langle O$ $|I\rangle\langle I|$ $|I\rangle\langle O$ Т Т $|O\rangle\langle I|$ $|O\rangle\langle I$ $|g\rangle\langle g|$ $|g\rangle\langle g|$ E_I E_o **Excited State Absorption** $\chi_{\scriptscriptstyle IOOI}$ $\chi_{{\scriptscriptstyle IIOI}}$ χ_{0001} χ_{OIOI} $|O\rangle\langle O|$ $|O\rangle\langle O|$ $|I\rangle\langle I|$ $|I\rangle\langle I|$ $|O\rangle\langle OI$ $I\rangle\langle II$ $|O\rangle\langle II$ $I\rangle\langle OI$ $|O\rangle\langle O$ $|I\rangle\langle I$ $|O\rangle\langle I|$ $|I\rangle\langle O$ Т Τ Т Т $|O\rangle\langle I|$ $|O\rangle\langle I|$ $|\overline{O}\rangle\langle\overline{I}$ $|O\rangle\langle I$ $|g\rangle\langle g|$ $|g\rangle\langle g|$ $|g\rangle\langle g|$ $|g\rangle\langle g|$ 7 $(2E_I - \Delta E_{II})$ $E_I - \Delta E_{IO}$ $E_I - \Delta E_{II}$ $-E_o$ $E_o - \Delta E_{IO}$



Figure A.1: Transient grating Feynman diagrams for quantum process tomography in doublewalled J-aggregate nanotubes. All possible Feynman diagrams pathways including secular and non-secular processes occurring during the delay time T. Green and blue arrows indicate interactions with energies E_I and E_O , respectively. Light green and light blue arrows indicate interactions with energies shifted by the biexciton binding energies $E_I - \Delta E$ and $E_O - \Delta E$, respectively. Yellow and purple arrows indicate interactions with energies $(2E_I - \Delta E_{II}) - E_O$ and $(2E_O - \Delta E_{OO}) - E_I$, respectively. ΔE_{xy} denotes the binding energy of the $|xy\rangle$ biexciton.

Appendix B

Two-Dimensional Rephasing Feynman Diagrams and Analysis for Cuprous Oxide

Appendix B enumerates all the possible Feynman diagrams for the two-dimensional rephasing spectrum of cuprous oxide, including transitions for pure exciton-polariton creation and destruction and phonon-assisted creation and destruction. Appendix B also includes the graphical maps of the two-dimensional rephasing signal intensity via Feynman diagram analysis.











Figure B.1: Two-dimensional rephasing Feynman diagrams in cuprous oxide. All possible orthoexciton pathways originating from the ground vibronic and phonon-populated ground electronic states. In the bra-ket notation, $|n, m\rangle$ represents a state occupied by n excitons and m phonons. Red, orange, and yellow arrows denote field interactions involving the changes in phonon occupation number by -1, 0, and +1, respectively.

(a)	a) Non-Overlapping Pathways							(b) Overlapping Pathways					
		Emission Energy (meV)								Emission Energy (meV)			
		2018		2032		2046				2004	2018	2032	2046
Excitation Energy (meV)	2046	A'	Α	B'	В	C'	С	on Energy (meV) 2032 2046	٥	A'	а	b	С
		-90	180	-18	36	-81	162		204	-99	178	-72	170
		-171*	342*	-18*	36*	-81*	162*				1/0	, _	-
	2032	D'	D	E'	E	F'	F		32	D'	d	е	F
		0	200	-10	20	-18	18		50	-1	191	3	20
		-18*	36*	-19*	38*	-18*	36*	itati	18	G'	g	h	I
	2018	G'	G	H'	н	ľ	I	EXc 20:	8	0	7	7	10
		0	0	0	0	0	0						
		0*	162*	-18*	36*	-171*	180*						

Figure B.2: Graphical displays for two-dimensional rephasing signal via Feynman diagram analysis. Graphical displays for non-overlapping (a) and overlapping (b) pathway strengths for all possible rephasing Feynman diagrams. * indicates pathways involving a phonon-populated ground electronic state. ' indicates excited state absorption pathways.

Appendix C

Two-Dimensional Two-Quantum Feynman Diagrams for Cuprous Oxide

Appendix C enumerates all the possible Feynman diagrams for the two-dimensional twoquantum spectrum of cuprous oxide, including transitions for pure exciton-polariton creation and destruction and phonon-assisted creation and destruction.





Ground Vibronic State y Feynman Diagrams

Ground Electronic State y Feynman Diagrams



Ground Vibronic State ε Feynman Diagrams

Ground Electronic State ε Feynman Diagrams



Figure C.1: Two-dimensional two-quantum Feynman diagrams in cuprous oxide. All possible orthoexciton pathways originating from the ground vibronic and phonon-populated ground electronic states. In the bra-ket notation, $|n, m\rangle$ represents a state occupied by n excitons and m phonons. Red, orange, and yellow arrows denote field interactions involving the changes in phonon occupation number by -1, 0, and +1, respectively.

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