Adsorption Structure and Mechanism of Styryl Phosphoric Acid at the Rutile–Water Interface

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Abstract: The microstructure and mechanism of styryl phosphoric acid (SPA) adsorbed at the rutile–water interface were investigated through zeta potential measurement, ultraviolet-visible spectrophotometry (UV-Vis), Fourier transform infrared spectroscopy (FT-IR) and X-ray photoelectron spectroscopy (XPS). The results of the zeta potential measurement illustrate that SPA is mainly electrostatically adsorbed on the rutile surface, and the adsorption process and result can be well fitted by the Stern-Grahame equation. The adsorption is severely affected by pH due to different species of SPA occurring in different pH solutions. The compound of P–O–Ti, with a structure of bidentate binuclear or bidentate mononuclear complexes, is formed after SPA is adsorbed on the rutile surface. SPA can be adsorbed on the rutile surface through the coordination of self-polymerization and bidentate mononuclear, which greatly increases the hydrophobicity of the rutile surface. Based on the above analysis and discussion, we proposed the adsorption model of SPA at the rutile–water interface, which was conducive to the modification and synthesis of a highly efficient flotation collector of the primary rutile ore.

Keywords: microstructure; electrostatic adsorption; chemical adsorption; surface hydroxylation

1. Introduction

Recently, the titanium industry has become more and more important and popular due to its extensive applications in medicine, navigation, aerospace, functional materials, and catalytic industries [1–6]. In the contemporary industrial system, titanium metallurgy sponge and titanium metal use rutile (TiO₂) and ilmenite (FeTiO₃) as raw materials. Compared with ilmenite, the titanium in rutile is relatively easy to utilize, with a natural advantage in quality. Therefore, many studies have begun to focus on rutile beneficiation [7–10]. The majority of rutile ores are refractory ores, the concentration of which is very difficult to gauge due to the fine grain sizes associated with the gangues, the complexity of the mineralogy, and the brittleness leading to it being easily over-ground [11]. Although the concentration of rutile ores is obtained by combining gravity, magnetic, and electrostatic separation techniques, flotation is one of the most efficient solutions to the issue [8].

In rutile flotation, sodium oleate, water-soluble petroleum sulfonate, styryl phosphoric acid (SPA), benzyl arsenic acid, alkyl imino bismethylene phosphoric acid and hydroximic acid are commonly used as collectors, and SPA is considered to be the most selective collector for rutile [12–18].

More and more mineral processing researchers are paying attention to the relationship between the adsorption of the collector at the mineral–water interface, and flotation recovery [19–21].
Many scholars have tried to explain the adsorption mechanism of collectors on mineral surfaces in the general formation of hydrogen bond adsorption, electrostatic adsorption and complex reaction adsorption [22–29].

It is of great significance to study the adsorption mechanism of collectors at the rutile–water interface, for the development and utilization of rutile [8,9,15]. Only when the adsorption mechanism of the collector on the mineral surface is clear enough can a flotation collector with low toxicity and high efficiency be designed by changing the functional group and carbon chain length. Madeley et al. [14] used anionic and cationic collectors as the collector for the flotation of natural rutile, indicating the rate of flotation was influenced by the length of the carbon chain associated with the collecting ions. Fuerstenau et al. [30] explained the adsorption phenomena of alkysulfonate at the rutile–water interface with the Stern-Grahame mode of the electrical double layer. Wang et al. [8] reported the flotation behavior and the mechanism of rutile with nonyl hydroxamic acid. The FT-IR results showed chemical adsorption identified on the rutile surface where a chelate of O, O-five-membered rings with Ti$^{4+}$ may form on the rutile surface. The density functional theory (DFT) calculation was used to study the adsorption behaviors of carboxylic acid at rutile TiO$_2$ (110) [31]. The calculation results showed that carboxylic acid preferred to be dissociatively adsorbed in a bridging bidentate configuration and induced significant surface relaxation at the adsorption sites, which also influenced other close surface atoms. Liu et al. [12] found that among the collectors of rutile flotation, such as benzyl arsenic acid, sodium oleate, sodium dodecyl sulphate, amino acids, diohosphonic acid, and styryl phosphoric acid (SPA), SPA was the most effective and aliphatic alcohol (e.g., octanol) was required to maintain the effectiveness. However, the mechanism of interaction between SPA and the rutile surface has rarely been reported. Peng [13] used angle resolution X-ray photoelectron spectroscopy (ARXPS) to study the effect of the interaction between SPA and octanol on rutile flotation and found that octanol associated with SPA pointed their long hydrocarbon chains towards bulk solution, while SPA was chemically adsorbed on the rutile surface. However, this literature does not fundamentally explain the mechanism of interaction between the collector and the rutile surface, especially at the molecular level of interpretation.

This paper attempts to explain the mechanism of interaction between SPA and the rutile surface using zeta potential measurement, UV-Vis, FT-IR, and XPS at the molecular level, and derived the adsorption model of SPA at the rutile–water interface. The present results are expected to be conducive to the understanding of the interaction mechanism between rutile and SPA and provide theoretical support for the development of a highly efficient flotation collector for primary rutile ores. However, model selection uncertainty exists in these models. In the future, we can assist DFT calculations and X-ray Absorption Fine Structure (XAFS) technology yet to be determined.

2. Materials and Methods

2.1. Materials and Reagents

The pure rutile samples used for the mechanism analysis were taken from Zaoyang Mine in Hubei, China, and then crushed, from which the rutile minerals were picked by hand. The handpicked rutile minerals were ground with a ceramic ball mill. The products after grinding were screened by 0.074 and 0.038 mm sieves, respectively. Those above 0.074 mm were returned to the ceramic ball mill for re-grinding, while those below 0.038 mm were stored as overcrushing samples. Particles from 0.038 to 0.074 mm were stored as experimental samples, and then purified by several magnetic separations, re-election impurity, and wet sieve. Finally, the particles were repeatedly cleaned using deionized water and dried at a low temperature as a flotation sample. The crystal phase composition was determined by X-ray diffraction (D8 Advance, Bruker, Karlsruhe, Germany), whose detailed processes and methods were previously described by Han [32]. The chemical composition of the rutile samples was measured by a wavelength dispersive X-ray fluorescence (XRF) spectrometer S4 Pioneer (PANalytical B.V., Almelo, The Netherlands). The XRD and XRF results are presented in Figure 1 and
Table 1. The former demonstrated that the rutile samples were mainly composed of rutile, and the Ti element only originated from the rutile mineral. Therefore, according to the element content of XRF shown in Table 1, the purity of the rutile mineral sample was calculated to be 93.8% with small impurities of 2.17% SiO₂ and 2.86% iron oxide.

Table 1. Multi-element analysis of the pure rutile mineral (%).

<table>
<thead>
<tr>
<th>Element</th>
<th>TiO₂</th>
<th>FeO</th>
<th>Fe₂O₃</th>
<th>SiO₂</th>
<th>CaO</th>
<th>MgO</th>
<th>Al₂O₃</th>
</tr>
</thead>
<tbody>
<tr>
<td>Content</td>
<td>93.80</td>
<td>1.53</td>
<td>1.33</td>
<td>2.17</td>
<td>0.17</td>
<td>0.31</td>
<td>0.99</td>
</tr>
</tbody>
</table>

Hydrochloric acid (analytical grade), sodium hydroxide (analytical grade) and potassium bromide (spectroscopically grade) were purchased from Sinopharm Chemical Reagent Co., Ltd. (Shanghai, China). Styryl phosphoric acid (SPA) with 68% purity was obtained from Zhuzhou Flotation Reagents, and then purified by recrystallization in methanol–water solvent in the laboratory. Its recrystallization product (95% purity) was used in all experiments. All aqueous solutions were prepared using Milli-Q water (Millipore deionized, 18.2 MΩ resistivity).

2.2. Methods

2.2.1. Zeta Potential Measurements

The zeta potential on the rutile surface was determined with a ZetaSizer 3000 Malvern Instrument (Malvern Instrument, Malvern, UK). The overcrushing samples were ground to less than 5 μm by agate grinding. The suspension was prepared by adding 20 ± 1 mg of TiO₂ to 50 mL of distilled, deionized water containing 10⁻³ mol/L KCl as a supporting electrolyte. The suspension was conditioned for 15 min, and the pH was measured and then adjusted using NaOH or HCl in the pH range of 2–11. Suspensions without or with 200 or 400 mg/L of SPA were placed on a rotating shaker for 30 min. The zeta potential was measured in accordance with the procedures described in the instrument manual. The reported results were the average of at least three full repeats of the experiment.

2.2.2. Adsorption Experiments

The SPA adsorption experiment was carried out using 2.000 ± 0.010 g of experimental samples. The aqueous solutions with 0–400 mg/L of SPA were added in sequence to the rutile samples in 100 mL
centrifuge tubes with 40 mL of liquid. Adsorption experiments were performed at the pH of 2.3 and 4.1, respectively. The samples were shaken with a centrifugal speed of 100 rpm for 12 h to achieve the equilibrium of adsorption determined with the preliminary test. Finally, the solid supernatants and sorbents were obtained from the slurry solution by filtering. The supernatants were analyzed through UV spectroscopy (UV-9100, Shimazu, Kyoto, Japan), and the adsorption capacity of SPA was calculated by the difference between the initial and equilibrium final concentrations, allowing us to determine the adsorption isotherms for SPA on the rutile mineral.

The experimental results were analyzed using Freundlich isotherms, as in the following Equation (1):

$$\log q_e = \log K_F + n_f \log C_e$$  \hspace{1cm} (1)

where $C_e$ is the equilibrium concentration of SPA in the liquid phase (mg/L), $q_e$ is the equilibrium concentration of SPA in the solid phase (mg/g), $K_F$ is the adsorption amount constant of the Freundlich model (mg$^{1/n}$·L$^{1/n}$·g$^{-1}$), and $n_f$ is the adsorption intensity constant of the Freundlich model.

The adsorption process of organic compounds on mineral particles is usually studied through Freundlich isotherms [33,34]. The Freundlich equation is an isotherm model that represents the adsorbent surface as heterogeneous [35]. Based on experimental data, the parameters of $K_F$ and $n_f$ were fitted according to the Freundlich isotherm model (Equation (1)) by means of the least square fitting technique, using nonlinear regression and the origin package software (Version 8.5).

2.2.3. FT-IR Spectroscopy of Rutile–SPA Complexes

FT-IR spectra of rutile, SPA and rutile–SPA complexes were recorded to examine the functional groups of rutile before and after adsorbing SPA. The experimental samples of 2.00 g were added into the 100 mL centrifuge tube with 40 mL of liquid, where the concentration of SPA was 8000 mg/L (400 mg/L × 20 times). By this time, the slurry pH was 2.3. Then, the samples were shaken at a centrifugal speed of 100 rpm for 30 min. FT-IR spectra were recorded over the region of 600–4000 cm$^{-1}$. The samples were examined in KBr pellets (3 mg/300 mg KBr). All samples were analyzed with a Lumex FTIR-08 spectrophotometer (PerkinElmer, Waltham, MA, USA).

2.2.4. XPS Measurements

The chemical compositions of the rutile surfaces were determined by XPS on a Thermo Scientific ESCALAB 250Xi (Thermo Fisher Scientific, Waltham, MA, USA), using an Al Kα X-ray source operated at 200 W with 20 eV pass energy. The sample preparation method was the same as that of the FT-IR spectra experiment (the SPA concentration was 400 mg/L, and the pH value was 2.3) with the vacuum pressure ranging from $10^{-9}$ to $10^{-8}$ Torr and a takeoff angle of 90°, which was used for collecting and analyzing data using the MultiPak spectrum software (Version 9.0) [36,37].

3. Results and Discussion

3.1. Zeta Potentials of the TiO$_2$ Surface

The adsorption of anions onto a mineral inevitably leads to a change in the surface charge [38,39]. Figure 2 shows the zeta potentials of the TiO$_2$ surface as a function of pH in the presence and absence of 200 or 400 mg/L of SPA. The obtained zeta potentials ranged from 20 mV to $-35$ mV. The isoelectric point (IEP) of the TiO$_2$ sample occurred at pH 4.5 where the net surface charge was zero, which is very consistent with the previously reported data [40].

The IEP of the TiO$_2$ surface shifted from 4.5 to 2.7 with the increase of the SPA concentration, and the zeta potentials showed negative values in the studied range of pH (from 2 to 9) even when the SPA total concentration in the solution was 400 mg/L. The shift of IEP to a lower pH value proved that anionic negatively charged surface complexes were formed [41]. Therefore, the results of the zeta potential measurement indicate that SPA is adsorbed on the TiO$_2$ surface in the form of negatively charged inner-sphere complexes. The formation of negatively charged SPA surface complexes can
be expected in advance because anionic C₆H₅C₂H₂PO₃H⁻ is the predominant SPA species in the pH range of 2 to 7.5, as shown in Figure 3. Therefore, the change of the isoelectric point on the surface of rutile in this experiment occurred fundamentally because the OH⁻ ion in the hydroxylation of the rutile surface was replaced by the C₆H₅C₂H₂PO₃H⁻ anionic. Thus, the process of the deprotonation of the rutile surface hydroxyl group is realized. Electrostatic adsorption may take place in the form of ion exchange that involves the displacement of similarly charged ionic species from the surfaces of the minerals [26].

Figure 2. Zeta potential of TiO₂ as a function of pH and total SPA concentration in 10⁻³ mol/L KCl solution.

The experiment was based on batch studies and showed negligible sorption of SPA in the alkaline solution, which was caused by a strong competition between the anionic C₆H₅C₂H₂PO₃H⁻ and OH⁻ on the TiO₂ surface in the alkaline solution. The result of this competition suggests that SPA could not be adsorbed on the TiO₂ surface in the alkaline solution, thus the form of SPA adsorption on the TiO₂ surface may be primarily electrostatic adsorption.

The adsorption density of collector ions, which was dominant with electrostatic adsorption, can be calculated using the Stern-Grahame Equation [42]:

\[ \Gamma = kC \exp \left( -\frac{\Delta G_{ads}^0}{RT} \right) \]  

(2)

where k (constant, and less than 1) is the adsorption coefficient; C is the collector concentration in aqueous solution; R is the gas constant; T is the absolute temperature; and \( \Delta G_{ads}^0 \) is the standard free energy of adsorption. \( \Delta G_{ads}^0 \) is estimated by the following Equation:

\[ \Delta G_{ads}^0 = ZF\Delta \zeta \]  

(3)

where Z is the valence of the adsorption ions (Z = 1 for SPA at pH 2.7); F is the Faraday constant; and \( \Delta \zeta \) is the difference value in zeta potential before and after the collector is adsorbed.

Based on the results in Figure 2, the adsorption density and standard free energy of SPA are shown in Table 2. \( \Delta G_{ads}^0 \) was a negative value in all conditions, indicating that SPA could be spontaneously adsorbed on the rutile surface in all experiments. The value of \( \Delta G_{ads}^0 \) represented the adsorption capacity of the collector on the mineral surface. The smaller the value, the stronger the adsorption capacity will be. Table 2 shows that the value (−2.82) of \( \Delta G_{ads}^0 \) for 400 mg/L of SPA adsorbed on the rutile surface was less than that (−1.99) for 200 mg/L of SPA. This reveals that the adsorption of 400 mg/L of SPA was easier than 200 mg/L of SPA. It is worth noting that the value of \( \Delta G_{ads}^0 \) for the adsorption difference between 200 mg/L of SPA and 400 mg/L of SPA was the smallest of all values,
which indicates that it may be difficult to continue to adsorb the SPA after a certain amount of SPA has already been adsorbed.

**Figure 3.** The percentage distribution of SPA species as a function of aqueous pH values, $K_\alpha 1 = 2.0$; $K_\alpha 2 = 7.1$ [43].

**Table 2.** The results of different parameters at pH 2.7 under various conditions.

<table>
<thead>
<tr>
<th>Condition</th>
<th>$\Delta \zeta$ (mV)</th>
<th>$\Gamma$ (mol/L)</th>
<th>$\Delta G_{ads}^o$ (kJ)</th>
</tr>
</thead>
<tbody>
<tr>
<td>$\Delta_1$</td>
<td>20.62 kC exp (0.000803)</td>
<td>-1.99</td>
<td></td>
</tr>
<tr>
<td>$\Delta_2$</td>
<td>29.25 kC exp (0.00114)</td>
<td>-2.82</td>
<td></td>
</tr>
<tr>
<td>$\Delta_3$</td>
<td>8.63 kC exp (0.000336)</td>
<td>-0.83</td>
<td></td>
</tr>
</tbody>
</table>

* $\Delta_1$: the adsorption of 200 mg/L SPA; $\Delta_2$: the adsorption of 400 mg/L SPA; $\Delta_3$: the adsorption difference between 200 and 400 mg/L SPA; $\zeta_1 = 19.35$ mV, $\zeta_2 = -1.27$ mV, $\zeta_3 = -9.91$ mV; $\Delta \zeta_1 = \zeta_1 - \zeta_2$, $\Delta \zeta_2 = \zeta_1 - \zeta_3$, $\Delta \zeta_3 = \zeta_2 - \zeta_3$.

### 3.2. Adsorption Isotherms

Figure 4 shows the adsorption isotherms and values of the Freundlich parameter ($K_F$ and $n_f$) for SPA adsorption on the TiO$_2$ surface. The adsorption isotherms for SPA on the TiO$_2$ surface were established in aqueous solution at pH 2.3 and 4.2 at 25 $^\circ$C. The adsorption data were well fitted to the Freundlich model ($r > 0.99$).

**Figure 4.** The Freundlich model fittings of SPA adsorption on the rutile surface.
From Figure 4, it can be seen that the change of pH values affects the isotherm curves, resulting in a change in the general shape of the isotherm. At pH 2.3, the adsorption amount constant ($K_F$) and the adsorption intensity constant ($n_f$) of the Freundlich isotherm model were 10.79 and 0.53, respectively, while they were 5.75 and 0.39 respectively at pH 4.1. The change of $K_F$ and $n_f$ indicates that the adsorption amount and the intensity increase with a decrease in pH, which is consistent with the results of the zeta potential measurement. Koopal et al. found that the adsorption process of ionic surfactants on the metal oxide mineral surface was typically characterized by four regions [44]. In this study, the adsorption process belonged to regions II and III. The difference between the two curves in Figure 4 revealed that pH played a very significant role in controlling the adsorption of ionic surfactants. Zhang et al. also reached a similar conclusion [27].

3.3. FT-IR

The presence of SPA adsorption on the rutile surface is shown in Figure 5, which presents the infrared spectrum of the SPA and the rutile before and after SPA adsorption. The adsorption of SPA is proven by the presence of the functional groups at 1157, 1452 and 2879 cm$^{-1}$, which correspond to the characteristic peaks of SPA in the infrared spectrum of rutile undergoing SPA adsorption. The peak of 1124 cm$^{-1}$ on the SPA curve is the characteristic of P=O, while the peaks of 993 and 957 cm$^{-1}$ are the characteristic of P–O [45,46]. However, it is only a peak of 1157 cm$^{-1}$ at about 957–1124 cm$^{-1}$ in the (SPA + rutile) curve. Compared with 1157 cm$^{-1}$, either 1124 or 957–1124 cm$^{-1}$ has a larger offset, which suggests a change in the chemical environment around the P and O atoms. This is because SPA chemically reacts with the Ti atom of the rutile surface, and the bond of P–O–Ti is produced. In the SPA curve, the stretching vibration band of the O–H···O group appears at 2854 cm$^{-1}$, which peak shifts to 2879 cm$^{-1}$ after SPA adsorption on the rutile surface. The peak of 2854 cm$^{-1}$ is relatively wide, while that of 2879 cm$^{-1}$ is relatively sharp, indicating that the polarity of the O–H group has changed. In observing the peak of 2854 cm$^{-1}$ in the SPA curve and the peak of 2879 cm$^{-1}$ in the (SPA + rutile) curve, it could be clearly seen that a hydrogen bond forms in the molecular structure of the SPA and rutile surface after SPA adsorption [47,48]. In the molecular structure of SPA, there are two hydroxyl groups that could form intramolecular hydrogen bonds and intermolecular hydrogen bonds between SPA and SPA molecular. However, there is a hydrogen atom in the hydroxyl group substituted by the Ti atom (forming P–O–Ti) on the rutile surface after SPA adsorption. The hydrogen bonds whose bond structure (P–O–H···O–Ti) is suggested on the rutile surface may be formed by a hydroxyl group in the SPA molecular with an oxygen atom.

![Figure 5. Infrared spectrum of SPA, rutile and rutile with SPA adsorbing treaded (at pH = 2.3).](image-url)
The molecules containing phosphate (PO(OH)$_2$) are characterized by the strong adsorption peak at 1613 cm$^{-1}$ in the SPA spectrum. The stretching vibration of the phenyl group (C$_6$H$_5$–) produces a characteristic intense band at 1452 cm$^{-1}$ [49]. The peak at 2361 cm$^{-1}$ is caused by the instrumental noise background. In addition, the characteristic adsorption peak of phosphate (PO(OH)$_2$) disappears in the (SPA + rutile) curve, which further proves that SPA chemically reacts with the Ti atom of the rutile surface, and metal–organic salt is produced in the adsorption process. The peak at 537 cm$^{-1}$ in the rutile curve and the peak at 536 cm$^{-1}$ in the (SPA + rutile) curve are characteristic of the adsorption of the titanium–oxygen octahedral [50,51], which suggests that the structure of the rutile surface (titanium–oxygen octahedral) did not change after SPA adsorption and thus the chemical bond of Ti–O–P may be formed between SPA and TiO$_2$. The hydroxylated surface is consistent with the titanium hydroxyl group (Ti–OH) in the analysis of the zeta potential and adsorption capacity.

3.4. XPS Analysis

Evidence provided by the FT-IR analysis suggests that the compound of P–O–Ti was formed after SPA adsorption on the rutile surface under acidic conditions. XPS analysis was carried out to further investigate the properties of the rutile surface and the behaviors of SPA adsorption on the rutile surface. The survey and high-resolution photoelectron spectra from the rutile samples before and after SPA adsorption are shown in Figure 6. Table 3 shows the binding energies and chemical shifts of the elements for rutile before and after SPA adsorption. The relative concentrations of the elements, as measured by XPS, are given in Table 4.

![Figure 6](image.png)

**Figure 6.** The survey scan and high-resolution X-ray photoelectron spectroscopy (XPS) of rutile before and after SPA adsorption.
Table 3. Binding energy of elements on the rutile surface before and after SPA adsorbing.

<table>
<thead>
<tr>
<th>Sample</th>
<th>Binging Energy (eV)</th>
<th>Chemical Shift (eV)</th>
</tr>
</thead>
<tbody>
<tr>
<td></td>
<td>C(1s)</td>
<td>O(1s)</td>
</tr>
<tr>
<td>Rutile</td>
<td>284.8</td>
<td>531.10</td>
</tr>
<tr>
<td>SPA</td>
<td>-</td>
<td>-</td>
</tr>
<tr>
<td>Rutile + SPA</td>
<td>284.9</td>
<td>531.49</td>
</tr>
</tbody>
</table>

Table 4. Relative content of elements on the rutile surface.

<table>
<thead>
<tr>
<th>Sample</th>
<th>Relative Contents (%)</th>
</tr>
</thead>
<tbody>
<tr>
<td></td>
<td>C(1s)</td>
</tr>
<tr>
<td>Rutile</td>
<td>25.01</td>
</tr>
<tr>
<td>Rutile + SPA</td>
<td>51.61</td>
</tr>
</tbody>
</table>

The apparent binding energies of the main Ti 2p3 of rutile and P 2p of organic phosphonate peaks at 458.54 and 134.35 ± 0.1 eV [52,53], respectively. In addition to the presence of elements that belong to minerals or agents, the presence of adventitious carbon on the mineral surface can also be observed. The relative composition of the natural rutile sample surface primarily includes C, O, and Ti. A P 2p peak is found in the spectra of the rutile samples after SPA adsorption, but single samples do not appear in rutile. This suggests that SPA could be adsorbed on the rutile surface in some formations, which is consistent with the results of the zeta potential measurement, UV-Vis, and FT-IR analysis. Table 4 shows that the concentration of P atoms was higher than that of Ti atoms on the rutile surface after SPA adsorption.

In rutile treated with SPA, the binding energies of Ti 2p and O 1s shifted by +0.43 eV and +0.39 eV, respectively. This result reveals a chemical interaction between the Ti or O atom and SPA on the rutile surface. The binding energy peaks of Ti 2p3 were 458.54 and 458.97 eV, respectively, which is characteristic of rutile titanium dioxide [54]. Therefore, the conclusion regarding the link between titanium atoms and phosphorus atoms through oxygen atoms (P–O–Ti) is correct. The binding energy shift of Ti 2p3 indicates that a change has taken place in the chemical environment around the Ti atoms. The chemical environment around the Ti atoms in the natural rutile is −Ti–O–Ti− or −Ti–O–H (after it is hydroxylated), but it changes into −Ti–O–P− after being treated with SPA. The electronegative value of the P atoms is greater than that of the Ti and H atoms, thus the electron cloud of the Ti and O atoms significantly shifts to the P atoms. The electron cloud density of the Ti atoms after being treated with SPA is greater than that of the untreated atoms, which explains why the binding energy of Ti 2p3 after being treated with SPA shifts to high energy [55,56]. Conversely, it could also mean that the binding energy of P 2p on the rutile surface shifts to low energy after SPA adsorbs on the rutile surface.

3.5. Establishment and Discussion of Adsorption Models

Based on the above experimental results and data, the adsorption molecular model of SPA at the rutile–water interface is established, as shown in Figures 7–9. In Figure 7, the molecular structures of SPA adsorption on the rutile surface are shown. He et al. [57] attempted to study the reaction pathway of arsenate adsorption at water−TiO2 interfaces based on the density function theory (DFT), finding that the bidentate binuclear adsorption structure was formed through a monodentate mononuclear metastable-equilibrium adsorption state. Figure 7a shows that SPA forms a bridge-type coordination system with the main adsorption structure of the bidentate binuclear. The bridge-type interaction has been reported in many studies [57]. Pena et al. [58] also found that arsenate formed a compound of (TiO)2AsO− on the surface of TiO2 in the form of bidentate binuclear coordination by extended X-ray absorption fine structure (EXAFS) analyses. However, Wang et al. [8] conjectured that ring-compounds of four or five members coordinating with the bidentate mononuclear were
formed in the study process of chelates on the surface of rutile adsorption. The molecular structure of O, O-four-membered ring-compounds with the bidentate mononuclear is shown in Figure 7b. The position of the hydrogen bond after SPA adsorption is shown in Figure 8. There may be three kinds of hydrogen bonds [26] in aqueous flotation systems: (1) the hydrogen bonds between the collector and water; (2) the hydrogen bonds between water and the mineral surface; and (3) the hydrogen bonds between collectors and the mineral surface, which are helpful for flotation. As the existence of hydrogen bonds is found by FT-IR spectroscopy on the rutile surface after SPA adsorption, and through the infrared testing process, there is no water involved [8, 59]. In our study, the hydrogen bonds between collectors and minerals are direct hydrogen bonds between collector functional groups and the surfaces of water solution minerals. In Figure 8a, the model of intramolecular hydrogen bonds is shown. As the phosphate-based polyhydroxy compounds can form intermolecular hydrogen bonds under certain conditions [60], we speculate as to the possibility of intermolecular hydrogen bonds after adsorption in Figure 8b. Figure 9 shows the adsorption models of one Ti atom and two SPA molecules, because the concentration of P atoms is higher than that of Ti atoms after SPA adsorption. Two models are modified on the basis of the bidentate mononuclear coordination. Due to the self-polymerization between phosphate groups, we present a model in Figure 9a. The adsorption methods in Figure 9 are helpful for the increased hydrophobicity of the rutile surface.

Figure 7. The conjecture of SPA adsorption on the surface of the rutile particle: (a) one SPA molecule binds two Ti atoms; (b) one SPA molecule binds to one atom.

Figure 8. The conjecture of the position of the hydrogen bond after the adsorption of SPA: (a) SPA in the formation of hydrogen bonding TiO₂; (b) SPA molecules to form hydrogen bonds.
These models in Figures 7–9 reveal the nature of SPA as the most efficient collector for rutile flotation [12]: (1) SPA can stably form the bidentate binuclear or bidentate mononuclear complexes on the surface of rutile; and (2) SPA can be adsorbed on the rutile surface in the form of coordination of the self-polymerization and bidentate mononuclear, greatly increasing the hydrophobicity of the rutile surface.

4. Conclusions

On the basis of the zeta potential and UV-Vis measurement, the formation of negatively charged surface complexes of C₆H₅C₂H₅PO₃H⁻ for SPA species adsorbed on the TiO₂ surface was determined. SPA was mainly electrostatically adsorbed on the rutile surface, whose adsorption process and result could be well fitted by the Stern-Grahame equation. The adsorption was severely affected by pH, due to different species of SPA occurring in different pH solutions.

Through FT-IR analysis, it was shown that the compound of P–O–Ti was formed after SPA was adsorbed on the rutile surface under acidic conditions. The hydrogen bond on the rutile surface after SPA adsorption may form by a hydroxyl group in SPA molecular with an oxygen atom on the rutile surface (P–O–H⋯O–Ti). The concentration of P atoms is higher than that of Ti atoms on the rutile surface after XPS analysis. This illustrates that SPA can be adsorbed on the rutile surface in the form of the coordination of self-polymerization and bidentate mononuclear, greatly increasing the hydrophobicity of the rutile surface.

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