Photodegradation of C.I. reactive red 2 in UV/TiO₂– In₂O₃–C–N and UV/TiO₂–In₂O₃–S–N systems

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Abstract In this study, the sol–gel method is utilized to produce $TiO₂–In₂O₃$ composite (Ti–In). The generated Ti–In was separately co-doped with C–N and S–N to form novel TiO_2 –In₂O₃–C–N (Ti–In–C–N) and TiO_2 –In₂O₃–S–N (Ti–In–S–N). The sources of N, C and S dopants were urea, powder activated carbon and $Na₂S₂O₃$. The dye C.I. reactive red 2 (RR2) was used as a model compound that was exposed to various composites and its removal by photocatalytic degradation and adsorption was measured. The percentage of anatase phase in Ti–In–C–N and Ti– In–S–N exceeded that in Ti–In; conversely, the particle diameter and band gap energy of Ti–In exceeded those of Ti–In–C–N and Ti–In–S–N. Bonds of Ti–O–N, Ti–N–O and Ti–O–C were detected in Ti–In–C–N and bonds of Ti–O–N, Ti–N–O and Ti–O–S were found in Ti–In–S–N. The pseudo-first order RR2 removal rate constants in UV/Ti–In, UV/Ti–In–C–N and UV/Ti–In–S–N were 0.43, 1.68 and 1.70 h^{-1} , in order. The specific oxygen uptake rates of the RR2 effluent in UV/Ti-In, UV/Ti–In–C–N and UV/Ti–In–S–N were determined to be 4.4, 7.3 and 5.3 mg O₂/g-MLVSS h. Ti-In-C-N and Ti-In-S-N were more effective in decolorizing and detoxifying RR2 than was their parent Ti–In.

Keywords $TiO_2 \cdot In_2O_3 \cdot C-N$ co-doping \cdot S-N co-doping \cdot Photodegradation

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Introduction

Advanced oxidation refers to a set of chemical treatment procedures for removing organic and inorganic materials from wastewater by oxidation. Photocatalysis using a nanosized metal oxide semiconductor is regarded as one of the most effective and simplest ways to mitigate environmental pollution that is caused by humans. Titanium dioxide (TiO_2) is currently considered to be the most promising photocatalyst because of its chemical and biological stability, high abundance and non-toxicity. The photocatalytic degradation of pollutants in water has been successfully performed using a suspension of $TiO₂$. The generally accepted mechanisms of the photodegradation of organics by $UV/TiO₂$ are as follows [[1\]](#page-12-0). Photogenerated holes are formed when $TiO₂$ particles are irradiated by UV light. Hydroxyl radicals are produced by the oxidation of OH^- or H_2O by these photogenerated holes, and are primarily responsible for the destruction of the organic species. Oxygen is used as an efficient electron trap to prevent the recombination of photogenerated electrons and holes. If the amount of available oxygen is limited, then the rapid recombination of photogenerated electrons and holes in $TiO₂$ reduces the efficiency of the photocatalytic reactions.

A different method for preparing photocatalytic $TiO₂$ efficiently involves coupling it with other semiconductor oxides, such as CdSe $[2]$ $[2]$, SnO₂ $[3, 4]$ $[3, 4]$ $[3, 4]$ $[3, 4]$, WO₃ $[4]$ $[4]$, and In_2O_3 [[5–7\]](#page-12-0). These composite systems demonstrate considerably enhanced photocatalytic performance in the degradation of pollutants, as the composite facilitates charge transfer and suppresses the recombination of electron–hole pairs in the photocatalysts systems. Coupling In_2O_3 with TiO_2 enables the transfer of the photogenerated electrons from the surface of In_2O_3 to the conduction band of TiO_2 because the band gap of In_2O_3 (2.64 eV) is lower than that of TiO₂ (2.95 eV) and the conduction band of In_2O_3 (-0.63 eV vs. NHE) is higher than that of TiO₂ $(-0.4 \text{ eV} \text{ vs. NHE})$. This process results in the effective capture of photogenerated electrons and retardation of the recombination of photogenerated electron–hole pairs [[7,](#page-12-0) [8\]](#page-12-0). Various studies have demonstrated that the photocatalytic activity of TiO_2 –In₂O₃ composite (Ti–In) exceeds that of In₂O₃ and TiO₂ [[5,](#page-12-0) [6](#page-12-0)].

Various metal and nonmetal ions have been doped into photocatalysts to reduce the recombination of photogenerated electrons and holes in them and to increase their absorption of light. However, metal ion-doped photocatalysts have some serious drawbacks, such as thermal instability and the metal centers acting as electron traps, reducing photocatalytic efficiency [\[9](#page-12-0)]. Sojic et al. [\[10](#page-13-0)] showed that nonmetals are more efficient dopants than metals. Accordingly, doping nonmetal ions into photocatalysts has recently attracted considerable attention. Nonmetal-doped forms of TiO₂, including C-doped $[11–13]$ $[11–13]$, N-doped $[10–12]$ $[10–12]$ $[10–12]$ and S-doped TiO₂ $[12, 14–16]$ $[12, 14–16]$ $[12, 14–16]$, has been found to exhibit superior photocatalytic activity because C, N and S atoms can effectively narrow the energy band gap of TiO₂. Moreover, C–N [[11\]](#page-13-0) and S–N codoped $TiO₂$ [\[17](#page-13-0), [18](#page-13-0)] have been investigated and co-doped $TiO₂$ has been found to have a higher photocatalytic activity than single-doped $TiO₂$ [[11,](#page-13-0) [18\]](#page-13-0).

Ma et al. [\[8](#page-12-0)] demonstrated that the photocatalytic activity of Pt/Ti–In exceeded that of Ti–In because the separation of the photogenerated electron–hole pairs in the former is more efficient, owing to the incorporation of In_2O_3 and Pt into the TiO₂

framework. Ag/Ti–In has been found to exhibit a similar enhancement [[19\]](#page-13-0). However, to the best of our knowledge, the photocatalytic activities of Ti–In that is co-doped with nonmetal dopants have not been examined. In this study, the Ti–In photocatalyst was synthesized via the sol–gel method, doped separately with C and S, and finally the C-doped and S-doped Ti–In were calcined with urea to form C–N co-doped and S–N co-doped Ti–In. To determine the photocatalytic activities of C– N-co-doped and S–N-co-doped Ti–In, their removal of a common azo dye, C.I. reactive red 2 (RR2), was studied. The objectives of this study are (i) to prepare novel C–N-co-doped Ti–In (abbreviated as Ti–In–C–N) and S–N-co-doped Ti–In (Ti–In–S–N), (ii) to characterize the surface properties of Ti–In–C–N and Ti–In–S– N, (iii) to determine changes associated with the adsorption of RR2 onto Ti–In–C–N and Ti–In–S–N, and (iv) to evaluate the photocatalytic activities of the doped Ti– In–C–N and Ti–In–S–N in decolorizing RR2, relative to that of their parent Ti–In.

Materials and methods

Materials

In this investigation, the sources of Ti and In are $TiCl₄$ (Acros) and $InCl₃$ (Alfa Aesar). The sources of N, C and S dopants were urea (Katayama), powder activated carbon (PAC) (Katayama) and $\text{Na}_2\text{S}_2\text{O}_3$ (Merck. The parent compound, RR2, was purchased from Sigma Aldrich. Its formula, molecular weight, and maximum light absorption wavelength were $C_{19}H_{10}Cl_2N_6Na_2O_7S_2$, 615 g/mol, and 538 nm. The pH of the solution was adjusted by adding 0.1 M HNO_3 (Merck) or NaOH (Merck). All compounds were used as received (without further purification), and all solutions were prepared using deionized water (Milli-Q) and reagent-grade chemicals.

Preparation of Ti–In, Ti–In–C–N and Ti–In–S–N

Ti–In was prepared with a Ti/In mole ratio of 54 using the sol–gel method, as previously described [\[7](#page-12-0)]. The molar ratios of Ti/C and Ti/S in Ti–In–C–N and Ti– In–S–N were both 7.5; the molar ratios of Ti/N in both Ti–In–C–N and Ti–In–S–N were both 0.07.

To prepare Ti–In–C–N and Ti–In–S–N, C-doped Ti–In (Ti–In–C) and S-doped Ti–In (Ti–In–S) were first produced. To prepare the Ti–In–C, PAC (0.184 g) was first mixed with HCl (20 mL; 0.65 M) and the mixture was placed in an ultrasonic bath for 10 min; then this solution was added InCl₃ (0.4805 g; 99.9 %) and TiCl₄ (13 mL; 98 %). The pH of the solution was adjusted to pH 8 by adding $NH₄OH$ solution (28 %), forming a solid precipitate. This precipitate was collected by filtration and washed repeatedly using distilled water; it was dried at 110° C for 24 h and then calcined at 450 °C for 2 h to yield the Ti–In–C. To prepare Ti–In–S, PAC was replaced with $Na₂S₂O₃$ (1.212 g) and the other steps were the same as those used to prepare Ti–In–C. To prepare Ti–In–C–N (or Ti–In–S–N), 2.5 g Ti–In– C (or Ti–In–S) and urea (12.5 g) were thoroughly mixed and the mixture was calcined at 450 \degree C for 2 h.

Characterization of Ti–In–C–N and Ti–In–S–N

The crystalline structures of various photocatalysts were analyzed by X-ray diffraction (XRD) (Bruker D8 SSS, Germany). The XRD patterns were recorded over 20 values of 10–90 $^{\circ}$, with an accelerating voltage and applied current of 40 kV and 30 mA, respectively. Diffuse reflectance UV–Vis spectra of the photocatalysts were obtained using a UV–Vis spectrophotometer (JAS.CO-V670, Japan) and these were used to calculate the band gap energies. Specific surface areas of the samples were determined by the BET method from the nitrogen adsorption isotherms that were obtained using a Micrometrics ASAP 2020 apparatus. X-ray photoelectron spectroscopy (XPS) measurements were made using a PHI Quantum 5000 XPS system (USA) with a monochromatic Al K_{α} source and a charge neutralizer.

Decolorization of RR2

Experiments were performed using 0.2 g/L of each photocatalyst, with $[RR2] = 20$ mg/L, at pH 3 and 25 °C. Decolorization experiments were conducted in a 3 L hollow cylindrical glass reactor, illuminated by an 8 W UV lamp (254 nm, Philips) with a light intensity of 1.12 W/m^2 that was placed inside a quartz tube. Adsorption experiments were performed in the dark. The reaction medium was continuously stirred at 300 rpm to maintain the suspension. Aliquots each with a volume of 10 mL were withdrawn from the reactor at intervals to monitor reaction progress. Following sampling, solids were separated by filtration using a $0.22 \mu m$ filter (Millipore), and the RR2 that remained in the filtrate was analyzed by measuring absorbance at 538 nm using a spectrophotometer (Hitachi U-5100, Japan).

Tests of specific oxygen uptake rate (SOUR)

The biodegradability of RR2 effluent was evaluated by performing a SOUR test. Activated sludge was collected from a sewage treatment plant in Kaohsiung. Before testing, the activated sludge was aerated for 24 h. The concentration of mixed liquor volatile suspended solids (MLVSS) in activated sludge was measured to be 3.5 g/L. The SOUR (mg O_2/g -MLVSS h) of aerobic microorganisms was measured for the RR2 effluent (after 180 min reaction and separated the suspended photocatalyst) to be the biodegradability index in the UV/Ti–In, UV/Ti–In–C–N and UV/Ti–In–S–N systems. SOURs were determined by dividing abatement rates in dissolved oxygen (DO) to MLVSS values.

Results and discussion

Characteristics of the surface of prepared photocatalysts

Fig. [1](#page-4-0) displays the XRD patterns of Ti–In, Ti–In–C–N and Ti–In–S–N. The anatase peaks were at 25.4 $^{\circ}$ and 48.2 $^{\circ}$. The peaks at 27.5 $^{\circ}$, 36.2 $^{\circ}$, 41.3 $^{\circ}$, 54.4 $^{\circ}$, and 69.2 $^{\circ}$ were attributable to rutile crystal phases. The peaks at 21.5° , 30.6° , 35.5° , 51.0° , and

Fig. 1 XRD patterns of Ti–In, Ti–In–C–N and Ti–In–S–N

Table 1 Parameters of surface properties for various photocatalysts

Photocatalysts	Anatase $(\%)$	Rutile $(\%)$	Diameter (nm)	Band gap (eV)	BET (m^2/g)	$k(h^{-1})$
$Ti-In$	30	70	57	2.97	35.7	0.43(0.999)
$Ti-In-C-N$	63	37	33	2.86	35.6	1.68 (0.988)
$Ti-In-S-N$	64	36	33	2.82	42.3	1.70 (0.982)

(): denoted the value of \mathbb{R}^2

60.7° were attributable to In₂O₃. No pattern included any In₂O₃-derived, N-derived, S-derived, or C-derived peaks.

The anatase content was determined from the integrated intensity of the anatase diffraction peak at $2\theta = 25.4^{\circ}$, I_A, and the rutile content was determine from the rutile diffraction peak at $2\theta = 27.5^{\circ}$, I_R, using Eq. 1 below [\[20](#page-13-0)]. From the XRD patterns, the crystalline size was calculated using the Scherrer equation, Eq. 2 [\[21](#page-13-0)]:

Anatase
$$
(\%) = \frac{1}{1 + 12.6 \frac{I_A}{I_R}} \times 100
$$
 (1)

$$
D = \frac{0.9\lambda}{\beta \cos \theta} \tag{2}
$$

Here, D represents the crystalline size (nm); λ is the wavelength of the X-rays (0.15418 nm); β is the line-width at the half maximum of the anatase peak at $2\theta = 25.4^{\circ}$ or of the rutile peak at $2\theta = 27.5^{\circ}$, and θ is the diffraction angle (°). Table 1 lists the surface properties of various photocatalysts. The surface areas followed the order Ti–In–S–N $>$ Ti–In \approx Ti–In–C–N. The percentage of anatase phase in Ti–In–C–N and Ti–In–S–N exceeded that in Ti–In. Clearly, C–N and S–N co-doping in Ti–In inhibited the conversion of $TiO₂$ from anatase to rutile. This retardation of phase transformation may be due to stabilization of the anatase phase by surrounding N ions via formation of Ti–O–N bonds. Anatase is regarded as the photoactive form of $TiO₂$, whereas rutile exhibits a low photocatalytic activity. Hurum et al. [\[22](#page-13-0)] proposed that electron-trapping sites are more stable in anatase than in rutile, so anatase exhibits greater photocatalytic activity. Accordingly, C–N and S–N co-doping in Ti–In may improve the photocatalytic activity of Ti–In. Additionally, C–N and S–N co-doping in Ti–In reduced the particle diameter, suggesting that the C–N and S–N co-doping retard the agglomeration of Ti–In crystals. Liu et al. [[16\]](#page-13-0) indicated that the doping with S effectively inhibited the growth of TiO₂ crystalls and Chen et al. [[11\]](#page-13-0) that doping with N had a similar effect. Li et al. [\[23](#page-13-0)] also demonstrated that carbon acts as a barrier that controls the growth of $TiO₂$ powder and prevents its agglomeration.

The band gap energy of Ti–In exceeded those of Ti–In–C–N and Ti–In–S–N (Table [1](#page-4-0)). N-doping was effective in reducing the band gap of $TiO₂$ by creating an isolated N_{2p} narrow band above the O_{2p} valence, by the incorporation of N atoms into the TiO₂ lattice [[24\]](#page-13-0). The reduction of the band gap of C-doped TiO₂ was attributed to the formation of Ti–O–C bonds [[25\]](#page-13-0). C–N co-doping generated the intragap localized state of C_{2p} and N_{2p} above the valence band gap of TiO₂. Therefore, electrons were promoted from the valence band to the conduction band, leaving holes behind [\[26](#page-13-0)]. The role of S in Ti–In–S–N was same as that of C in Ti– In–C–N; therefore, co-doping of C–N and S–N in Ti–In reduced the band gap energy.

XPS was utilized to determine the chemical compositions of the photocatalysts. Fig. [2](#page-6-0)a–e display the XPS spectra of Ti_{2p} , In_{3d} , O_{1s} , C_{1s} , and N_{1s} for Ti–In–C–N. The $Ti_{2p3/2}$ and $Ti_{2p1/2}$ spin-orbital-splitting photoelectrons were observed at binding energies of 455.7–458.8 and 462.1–464.5 eV [[26–28\]](#page-13-0). The binding energy of the $Ti_{2p3/2}$ and $Ti_{2p1/2}$ spin-orbital-splitting photoelectrons in Ti–In–C–N was 458.2 and 463.9 eV (Fig. [2\)](#page-6-0). These binding energies are close to those reported for Ti^{4+} in TiO₂ [[29\]](#page-13-0). The In_{3d5/2} and In_{3d3/2} spin-orbital splitting photoelectrons were observed at binding energies of 443.9–445 [\[19](#page-13-0), [28](#page-13-0), [30](#page-13-0), [31](#page-13-0)] and 451.5–452.0 eV [[19,](#page-13-0) [28](#page-13-0), [30](#page-13-0)]. The binding energy of the $In_{3d5/2}$ and $In_{3d3/2}$ spin-orbital-splitting photoelectrons of Ti–In–C–N was 444.3 and 451.9 eV (Fig. [2](#page-6-0)b). Peaks at 528.5–529.5 [\[26](#page-13-0), [28](#page-13-0), [32\]](#page-13-0) and 529.6–531.0 eV [\[27](#page-13-0), [30,](#page-13-0) [31](#page-13-0)] were assigned to O_{1s} ; the former was associated with Ti–O and the latter was attributable to In–O. Peaks at $531.4-531.9$ eV indicated other oxygen species, possibly belonging to OH^- on the surface of the photocatalyst [[32,](#page-13-0) [33](#page-13-0)]. The binding energy of Ti–O, In–O, and OH^- in Ti–In–C–N was 529.4, 529.7, and 531.6 eV (Fig. [2c](#page-6-0)). The binding energies of C_{1s} , which were 284.8, 285.4 and 288 eV, were attributed to C–C, C–O [[34](#page-13-0)], and Ti–O–C [[27\]](#page-13-0) bonds (Fig. [2d](#page-6-0)). The 284.8 eV peak is due to the carbon-containing species adsorbed on the surface, which is usually ascribed to the adventitious carbon or carbon residues from the organic precursor [\[12](#page-13-0)]. Kang et al. [[35\]](#page-13-0) assigned the binding energy at 282 eV to the Ti–C bond that was formed by the substitution of carbon at an oxygen site in $TiO₂$; they assigned the binding energy at 288 eV to the presence of carbonate species that were formed by the substitution of carbon at the Ti site in $TiO₂$. The prepared Ti–In–C–N formed C–C and C–O bonds, but no Ti–C bonds. Chen et al. $[11]$ $[11]$ reported that carbonate species that formed on the TiO₂

Fig. 2 XPS spectra of Ti-In-C-N. a Ti_{2p}, b In_{3d}, c O_{1s}, d C_{1s}, e N_{1s}

Fig. 2 continued

surface served as a photosensitizer, like an organic dye, promoting the absorption of $TiO₂$.

Fig. [3](#page-8-0)a–e present the XPS spectra of Ti_{2p} , In_{3d} , O_{1s} , N_{1s} , and S_{2p} for Ti–In–S–N. The XPS spectra of Ti_{2p} , In_{3d} and O_{1s} in Ti–In–S–N (Fig. [3a](#page-8-0)–c all revealed similar binding energies to those in Ti–In–C–N (Fig. [2a](#page-6-0)–c). Binding energies of 397.0–397.4 [[36–](#page-13-0)[38\]](#page-14-0) and 398.6 eV [[36,](#page-13-0) [37\]](#page-14-0) in the N_{1s} region were assigned to Ti–N–O; the binding energy at 399–400.7 eV was attributed to the Ti–O–N bond [\[28](#page-13-0), [31](#page-13-0)]. For Ti–In–C–N, Ti–N–O and Ti–O–N were obtained at the binding energy of 397.3 and 399.6 eV (Fig. [2](#page-6-0)e). Moreover, for Ti–In–S–N, Ti–N–O and Ti–O–N were obtained at the binding energy of 398.7 and 400 eV (Fig. [3](#page-8-0)d). The XPS spectra of Ti_{2p} and In_{3d} for Ti–In–C–N and Ti–In–S–N were similar to those for Ti–In [\[7](#page-12-0)] and Ti–In–N $[39]$ $[39]$; moreover, the XPS spectra of O_{1s} and N_{1s} for Ti–In–C–N and Ti–In–S–N were similar to those obtained for Ti–In–N by Wu et al. [[39\]](#page-14-0) (data not

Fig. 3 XPS spectra of Ti-In-S-N. a Ti_{2p}, b In_{3d}, c O_{1s}, d N_{1s}, e S_{2p}

Fig. 3 continued

shown). Zhang et al. [\[40](#page-14-0)] indicated the post-nitridation of $TiO₂$ by NH₃ takes only place on the surface layer and sublayer of $TiO₂$ and maintains the original crystal phase of TiO₂ below 500 °C; accordingly, no crystalline phase of Ti–N was obtained. The present study identified two crystalline phases, Ti–O–N and Ti–N–O, which may be attributable to the use of urea rather than $NH₃$ as the precursor of N. Asahi et al. [[24\]](#page-13-0) utilized urea as a dopant and concluded that nitrogen atoms substituted at the lattice oxygen sites, forming an isolated narrow band above the valence band and narrowing the band gap. Notably, the preparation procedure varied among the studies cited above, and this variation may explain the various XPS observations.

The S atoms herein are in the S^{6+} state, yielding a peak at 168.2–169 eV [[14,](#page-13-0) [16,](#page-13-0) [41](#page-14-0)]. The binding energy of S_{2p} in Ti–In–S–N is 168.7 eV (Fig. [3e](#page-8-0)). When S atoms

Fig. 4 RR2 removal in Ti–In, Ti–In–C–N and Ti–In–S–N systems via adsorption and photodegradation $([photocatalyst] = 0.2 g/L, [RR2] = 20 mg/L, pH 3)$

replace O atoms on the TiO₂ surfaces, no peak in the range $160-163$ eV, which corresponds to the Ti–S bond, is obtained [\[14](#page-13-0)]. Umebayashi et al. [[42\]](#page-14-0) suggested that sulfur was doped as an anion and replaced the lattice oxygen in $TiO₂$ to form Ti–S bond, conversely, Ohno et al. [[41\]](#page-14-0) reported that S atoms were incorporated as cations and replaced Ti ions in the S-doped TiO₂. Ho et al. $[15]$ $[15]$ suggested that the oxidation state of the S-dopant depends on the preparation route. Anionic S doping may be difficult to perform because S^{2-} (0.17 nm) has a significantly larger ionic radius than does Q^{2-} (0.122 nm). Hence, the substitution of Ti^{4+} by S^{6+} is chemically more favorable than replacing Q^{2-} with S^{2-} [\[14](#page-13-0)], so Ti–In–S–N becomes positively charged. Therefore, the efficiency of adsorption of RR2 by Ti– In–S–N may exceed those of Ti–In–C–N and Ti–In.

Li et al. $[18]$ $[18]$ found that the absorption spectra of S–N co-doped TiO₂ exhibited a red shift in the adsorption edge. This shift was attributed to the fact that calcination induces the doping of S and N atoms into the lattice of $TiO₂$, narrowing the band gap. Wang and Lim $[26]$ $[26]$ observed a red shift of the absorption edge of C–N codoped $TiO₂$ owing to the coexistence of Ti–C bonds, carbonate and oxynitride species in the $TiO₂$ lattice. The experimental results for Ti–In–C–N and Ti–In–S–N herein were similar to those for C–N co-doped $TiO₂$ and S–N co-doped $TiO₂$.

Photocatalytic properties of Ti–In, Ti–In–C–N and Ti–In–S–N

At pH 3, no significant decolorization occurred during direct photolysis $(\leq 5 \%)$ (data not shown herein). Hence, the disappearance of RR2 was attributed to adsorption and/or photodegradation reactions. Fig. 4 presents the decolorization of RR2 by adsorption and photodegradation in different systems. After 180 min, the

Fig. 5 DO uptake tests for RR2 effluent in in UV/Ti–In, UV/Ti–In–C–N and UV/Ti–In–S–N systems $([photocatalyst] = 0.2 g/L, [RR2] = 20 mg/L, pH 3)$

adsorptions of RR2 (in the dark) by Ti–In, Ti–In–C–N and Ti–In–S–N were 34, 20 and 43 %, while the total removals of RR2 (adsorption and photodegradation under illumination) were 78, 97 and 99 %. Therefore, the net removal of RR2 that was attributable to photodegradation by various photocatalysts followed the order Ti– In–C–N $>$ Ti–In–S–N $>$ Ti–In. The order of RR2 adsorption by these photocatalysts was Ti–In–S–N $>$ Ti–In $>$ Ti–In–C–N, which agrees with the order of the surface areas of these photocatalysts. The removal of RR2 in all experiments followed pseudo-first order kinetics, as has been reported in various studies of RR2 decolorization [[7,](#page-12-0) [13](#page-13-0)]. Table [1](#page-4-0) lists the pseudo-first order RR2 removal rate constants (k), which follow the order $Ti-In-S-N > Ti-In-C-N > Ti-In$. The k values in UV/Ti–In, UV/Ti–In–N [\[39](#page-14-0)], UV/Ti–In–C–N, and UV/Ti–In–S–N were 0.43, 1.60, 1.68 and 1.70 h^{-1} . Clearly, the promotion of photocatalytic activity of Ti–In by co-doping with C–N or S–N was greater than that by N doping.

This study suggested that N atoms could form Ti–N–O and Ti–O–N, while C atoms could form a mixed layer deposited PAC and Ti–O–C bond at the surface of Ti–In–C–N particles. Furthermore, S atoms were incorporated as cations and replaced Ti ions in Ti–In–S–N. Based on the experimental analysis, C, S and N were successfully doped into Ti–In. The improved photocatalytic activity can be evidenced by an increase in either the number of Ti^{4+} defects or the number oxygen vacancies upon doping, the form of the impurity states, which inhibit the recombination of photogenerated hole and electron pairs, reduced band gap energy and increased lifetime of holes and electrons [[43\]](#page-14-0). Yu et al. [\[17](#page-13-0)] suggested that the high activity of N–S co-doped $TiO₂$ can be attributed to the red shift of the adsorption edge, a large surface area and the two phase structures of N–S co-doped TiO2. Large surface areas facilitate contact between holes (or electrons) and the pollutant molecules, improving photocatalytic activity.

Fig. [5](#page-11-0) plots the DO profiles that were obtained by respirometric measurements of the RR2 effluent in different systems. The SOURs for the RR2 effluent (after 180 min reaction) in UV/Ti–In, UV/Ti–In–C–N and UV/Ti–In–S–N were determined to be 4.4, 7.3 and 5.3 mg O_2/g -MLVSS h. The C–N and S–N co-doped Ti–In improved the biodegradation of RR2 effluent. The enhancement of biodegradation by Ti–In–C–N exceeded that by Ti–In–S–N. The experimental results revealed that Ti–In–C–N effectively decolorized and detoxified RR2.

Conclusions

Novel Ti–In–C–N and Ti–In–S–N photocatalysts were fabricated in this study. C–N and S–N-co-doping in Ti–In inhibited the transformation of $TiO₂$ from anatase to rutile and particle agglomeration. The bang gap energy and particle diameter of Ti– In–C–N and Ti–In–S–N were smaller than those of Ti–In. The RR2 removal rates followed the order Ti–In–S–N $>$ Ti–In–C–N $>$ Ti–In. In this investigation, C, S and N were successfully doped into Ti–In, and the resulting improved photocatalytic activity was attributed to an increase in either the number of Ti^{4+} defects or the number of oxygen vacancies, which inhibited the recombination of photogenerated hole and electron pairs. Ti–In–C–N and Ti–In–S–N effectively decolorized and detoxified RR2.

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