

# Simultaneous and temporal autoregressive network models

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## Abstract

While logistic regression models are easily accessible to researchers, when applied to network data there are unrealistic assumptions made about the dependence structure of the data. For temporal networks measured in discrete time, recent work has made good advances (Almquist & Butts, 2014), but there is still the assumption that the dyads are conditionally independent given the edge histories. This assumption can be quite strong and is sometimes difficult to justify. If time steps are rather large, one would typically expect not only the existence of temporal dependencies among the dyads across observed time points but also the existence of simultaneous dependencies affecting how the dyads of the network co-evolve. We propose a general observation-driven model for dynamic networks that overcomes this problem by modeling both the mean and the covariance structures as functions of the edge histories using a flexible autoregressive approach. This approach can be shown to fit into a generalized linear mixed model framework. We propose a visualization method that provides evidence concerning the existence of simultaneous dependence. We describe a simulation study to determine the method's performance in the presence and absence of simultaneous dependence, and we analyze both a proximity network from conference attendees and a world trade network. We also use this last data set to illustrate how simultaneous dependencies become more prominent as the time intervals become coarser.

**Keywords:** *dependence structures, dynamic networks, generalized linear mixed models, multivariate probit, observation-driven model*

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## 1 Introduction

Co-occurrence data involves observing a set of interactions, or edges, between a set of actors. The observed edge set and actor set together form a network object. Such networks arise in multitudinous contexts, and the analysis of network objects has been of extreme importance to scientists in a wide range of fields. In particular, the analysis of network dynamics is an extremely interesting and often difficult area to work in, as temporal dependencies are added to an already complex network dependence structure.

Several classes of models for temporally measured, or dynamic, networks have been proposed, mostly over the last two decades. Each of these classes comes with pros and cons, as one would expect. The network literature is vast even for dynamic networks, and so we only touch on a few of the key classes of models before presenting our proposed approach.

39 Modeling dynamic networks using continuous-time Markov processes has a long  
40 history beginning with Holland & Leinhardt (1977) and continuing with several  
41 other works (e.g., Wasserman, 1980; Leenders, 1995). A very impactful work  
42 continuing the adoption of continuous-time Markov processes is the stochastic  
43 actor-oriented model (Snijders, 1996), which has since seen much methodological  
44 and software development (Ripley *et al.*, 2013). In this framework, each actor forms  
45 a new edge or breaks an existing edge in order to maximize that actor's so-called  
46 objective function. This function can represent homophily on attributes or structures  
47 of the network itself, such as transitivity and reciprocity. This class of models has  
48 been very popular and useful, and allows for wide flexibility in constructing the  
49 objective function.

50 Another popular class of models used for static networks is the exponential  
51 random graph (ERG) models, proposed by Frank & Strauss (1986) and developed  
52 further in countless works. The ERG family of models was extended to dynamic  
53 networks by Robins & Pattison (2001), and later extended by Hanneke *et al.* (2010)  
54 and others. The temporal ERG model, or TERG model, in contrast to the stochastic  
55 actor-oriented model, assumes the network data to be generated according to  
56 a discrete time Markov process. The general idea in these ERG models is to  
57 put the probabilistic structure of the observed networks in terms of functions of  
58 sufficient statistics. These statistics often correspond to a count of some topological  
59 feature, such as triangles or  $k$ -stars. The TERGM is quite flexible in the sufficient  
60 statistics that can be included in the model, is parsimonious, and can handle  
61 complex dependencies in the network. Similar in spirit is the Separable TERGM  
62 (Krivitsky & Handcock, 2014), where both the formation and dissolution process  
63 are modeled. Unfortunately, there are a variety of problems that arise with these  
64 types of ERG models. There is the intractable normalizing constant that must be  
65 approximated, as well as degeneracy issues, or non-existence of the maximum  
66 likelihood estimators. See, e.g., Okabayashi (2011) and Jin & Liang (2013) for  
67 more on this, as well as Hummel *et al.* (2012) for remedies to some of these  
68 problems.

69 Stochastic blockmodels (Holland *et al.*, 1983; Wang & Wong, 1987;  
70 Snijders & Nowicki, 1997) have been one of the most widely used and studied  
71 class of models for networks. The mixed membership blockmodel (Airoldi *et al.*,  
72 2008) was extended for dynamic networks by Xing *et al.* (2010). While quite useful,  
73 blockmodels suffer from an inability to capture network dependencies induced by  
74 complex features such as transitivity or reciprocity.

75 A large number of models fall into the class of latent space models. These models  
76 originated with Hoff *et al.* (2002) for static networks, and expanded in a variety of  
77 ways (see, e.g., Handcock *et al.*, 2007; Krivitsky *et al.*, 2009). These models were then  
78 extended to the dynamic context by Sarkar & Moore (2005), Durante & Dunson  
79 (2014), and Sewell & Chen (2015). Scalability remains an issue with latent space  
80 models, though some attempts have been made to alleviate this (Raftery *et al.*,  
81 2012; Salter-Townshend & Murphy, 2013), and determining the dimensionality of  
82 the latent space has attracted relatively little serious work, the main exception being  
83 work done by Durante & Dunson (2014).

84 Our proposed work builds off of the logistic network regression models proposed  
85 by Almquist & Butts (2013, 2014). This model provides a simple yet flexible

86 framework for capturing the temporal dependency by modeling the mean as  
87 a function of sufficient statistics constructed from previous observations of the  
88 network. Their model has distinct advantages such as scalability, flexibility, and  
89 easy accessibility to anyone familiar with generalized linear models. The authors  
90 derive this model from the TERGM based on a clear set of assumptions. The most  
91 controversial of these is that the network dyads are conditionally independent given  
92 the network history. The problem is that the simultaneous dependence is ignored, i.e.,  
93 the dependence between the co-evolving dyads. These simultaneous dependencies  
94 play an important role in the evolution of the network, especially as the intervals  
95 at which the network is observed increase (Lerner *et al.*, 2013). It is well known  
96 that ignoring extra variation in the data can, in contexts similar to our own, lead  
97 to inconsistent estimation and attenuated estimates of the parameters (Demidenko,  
98 2013). Thus ignoring simultaneous dependence in the data will in many cases lead to  
99 poor estimation; we shall demonstrate this analytically in Section 2.3 and empirically  
100 in Section 6.

101 Cox (1981) used the terms “parameter driven” and “observation driven” models  
102 to describe two approaches for modeling binary time series data. In the context of  
103 dynamic network analysis, we can think of the latent space approach as the analog to  
104 parameter-driven models, where the temporal dependencies of the network are driven  
105 through some latent variables evolving through, say, a Markov process. Our proposed  
106 model follows what may be considered an observation-driven approach, where  
107 both the simultaneous and temporal dependencies are driven by some functions of  
108 the lagged observed networks. More specifically, our proposed approach captures  
109 temporal dependence through modeling the mean as a function of lagged networks  
110 and similarly captures the simultaneous dependence through modeling the covariance  
111 as a function of lagged networks.

112 An important motivation for this work was accessibility to appropriate network  
113 methodology for those without extensive statistical background. We believe that  
114 those familiar with generalized linear mixed models (GLMMs) (see Section 4)  
115 should be able to easily understand and utilize our proposed approach, and  
116 software will be made available on the author’s website to further facilitate  
117 accessibility. While using a familiar framework, we account for both temporal  
118 and simultaneous dependence, thus avoiding the adverse inferential impacts  
119 that we otherwise would expect to occur by ignoring these two sources of  
120 variation.

121 In Section 2, we present our proposed methodology, as well as some suggestions  
122 for appropriately choosing the mean and covariance functions. In Section 3, we  
123 describe our approach to estimation, with the details and selected proofs given in the  
124 appendix. Section 4 generalizes our approach by fitting our method into the familiar  
125 GLMM framework. In Section 5, we describe a visualization approach to evaluating  
126 the evidence regarding the existence and impact of simultaneous dependence in the  
127 data. In Section 6, we present a simulation study that examines the performance of  
128 our model in the presence and absence of simultaneous dependencies. In Section 7, we  
129 analyze two real data sets, illustrating the utility of our method and the importance  
130 of accounting for simultaneous dependence in real data, as well as illustrating  
131 how simultaneous dependence becomes more prominent as time intervals become  
132 coarser.

## 2 Methodology

### 2.1 Context and notation

We assume that we have  $n$  objects, or *actors*, each of which may have some interactions or relationships with the other actors. If such an interaction/relationship exists between actors  $i$  and  $j$ , we say there is an *edge* between them. We assume that the set of actors are constant over time, though the edges themselves may exist during any subset of all possible time points. Here, we assume the data are collected at discrete time points. Collectively, the set of actors and the time-varying set of edges define the dynamic network. The data obtained can then be represented by a three-dimensional tensor, or equivalently a sequence of adjacency matrices, where each adjacency matrix, denoted as  $A_t$ ,  $t = 0, 1, \dots, T$ , is an  $n \times n$  matrix corresponding to the edges that exist at time  $t$ . That is, the  $(i, j)$ -th entry of  $A_t$ ,  $A_{ijt}$ , equals one if there is an edge from  $i$  to  $j$  at time  $t$  and zero otherwise. The diagonal entries of each adjacency matrix hold no meaning unless so-called self-loops are allowed, that is, an actor may send an edge to itself. For the purposes of clarity in our exposition, we will assume in Section 2 that such self-loops are allowed as this helps facilitate the mathematical description of the model and its properties; it is trivial to translate the presented model to the context of no self-loops. However, because (1) self-loops are relatively rare in practice, and (2) the derivations of our estimation algorithm requires additional non-trivial steps when self-loops are not allowed, the derivations provided in our appendices assume the diagonal elements of the  $A_t$ 's are meaningless. Additionally, the data in Sections 6 and 7 do not have self-loops.

We also assume there exist some exogenous covariate information with which we would like to explain or predict the edge probabilities. These covariates may be static (e.g., race or gender) or time-varying (e.g., income or marital status). In the remainder of the paper, we will treat the covariates as though they are time-varying with the understanding that static covariates may be treated as such simply by replicating them from one time point to the next. We denote the dyadic covariate information by the  $n \times n$  matrices  $X_{\ell t}$ ,  $\ell = 1, \dots, p_1$ ,  $t = 1, \dots, T$ . For notational convenience, we will denote a linear combination of equal sized matrices as  $\langle \boldsymbol{\beta}, \mathcal{X}_t \rangle := \sum_{\ell=1}^{p_1} \beta_{\ell} X_{\ell t}$ , where  $\boldsymbol{\beta} = (\beta_1, \dots, \beta_{p_1})$  and  $\mathcal{X}_t$  is a 3-dimensional array whose  $\ell$ th slice is  $X_{\ell t}$ .

As will be seen shortly, we shall be focusing on covariance structures, and hence it is natural to implement a probit type model for our binary dyadic data (although we will generalize the work in Section 4). We thus assume that there are some underlying matrices of normal random variables  $A_t^*$  that directly correspond to  $A_t$  via the surjective function  $A_{ijt} = \mathbf{1}_{\{A_{ijt}^* > 0\}}$ .

### 2.2 Observation-driven model

The proposed model is an observation-driven approach, rather than parameter-driven. That is, we may write the conditional mean of  $A_t^*$  as a function of  $A_0, \dots, A_{t-1}$  rather than as a function of some unobservable noise process. Observation-driven approaches for temporal binary data have been well studied in simpler contexts. While some complicated mean functions have been proposed (e.g., Shephard, 1995),

often it is the simple and intuitive

$$\mathbb{E}(A_{ijt}^* | A_{ij(t-1)}, A_{ij(t-2)}, \dots) = \sum_{\ell=1}^{p_1} \beta_{\ell} X_{\ell t}[i, j] + \sum_{\ell=1}^{p_2} \theta_{\ell} A_{ij(t-\ell)},$$

(e.g., Cox, 1981; Zeger & Qaqish, 1988), where  $X[i, j]$  is the  $(i, j)$ -th entry of the matrix  $X$ . However, this simplistic mean function is insufficient for complex network objects. With this in mind, we will allow the second term of the mean of  $A_t^*$  to be  $\langle \boldsymbol{\theta}, \mathcal{G}_t \rangle := \langle \boldsymbol{\theta}, \mathcal{G}(A_{t-1}, A_{t-2}, \dots) \rangle$ , where  $\boldsymbol{\theta} = (\theta_1, \dots, \theta_{p_2})$ , and  $\mathcal{G}_t$  maps the previous adjacency matrices onto the space of  $n \times n \times p_2$  tensors, i.e.,  $\mathcal{G}_t$  uses the previous adjacency matrices to construct  $p_2$  new  $n \times n$  matrices.

Note that  $p_2$  does not refer to the number of lagged time points as in the simple binary time series model, but rather can encompass a number of salient features of the previous adjacency matrices, such as stability, reciprocity, or transitivity. As a simple example, if we include stability and reciprocity for up to a lag of two time points, then  $p_2 = 4$  and the slices of  $\mathcal{G}_t$  are  $A_{t-1}$ ,  $A'_{t-1}$ ,  $A_{t-2}$ , and  $A'_{t-2}$ . These  $p_2$  covariates involving functions of the lagged network can thus be used in sophisticated ways to explain the temporal dependencies, i.e., the dependence between  $A_{ijt}$  and  $A_{k\ell s}$ ,  $t \neq s$ . For examples of other ways to construct  $\mathcal{G}_t$ , see Table 1 or the appendices of Almquist & Butts (2014).

Networks are complex objects, however, and attempting to capture all dependencies through the mean structure alone is insufficient, particularly as the intervals between time points grow larger. One would typically expect not only the existence of *temporal* dependencies through which the network at varying time points are dependent, but also *simultaneous* dependencies that dictate how the dyads of the network co-evolve. Thus, we should be quite concerned with appropriately modeling the second moments of the  $A_{ijt}^*$ 's.

With this motivation in mind, we begin with the following multivariate probit model. Let  $\mathcal{A}_t$  be equal to  $\text{vec}(A_t^*)$ . Then set

$$\mathbb{E}(A_t^* | A_{t-1}, A_{t-2}, \dots) = \langle \boldsymbol{\beta}, \mathcal{X}_t \rangle + \langle \boldsymbol{\theta}, \mathcal{G}_t \rangle \tag{1}$$

$$\text{Cov}(\mathcal{A}_t) = \Sigma_{A^*, t}. \tag{2}$$

Note that  $\Sigma_{A^*, t}$  determines the covariance structure among the  $n^2$  dyads, and hence has  $\mathcal{O}(n^4)$  parameters. Clearly, it would not be possible to estimate such an unconstrained  $\Sigma_{A^*, t}$  outside of the context of small  $n$  large  $T$ , nor is this unconstrained covariance structure what one would expect to see in reality. Going to the extreme of constraining  $\Sigma_{A^*, t}$  to be the identity matrix (and thus ignoring simultaneous dependence entirely) leads to the model presented in Almquist & Butts (2014), and hence what is presented here can be thought of as an alternative generalization of their methods (the TERGM is the original motivation for and generalization of their approach).

### 2.3 Ignoring simultaneous dependencies

Here, we make a short note on estimation errors associated with ignoring existing variability in the data. Demidenko (2013) gives a short discussion on these types of issues with regard to GLMMs (see chapter 7). For our context, suppose we may

215 write the normal random variables  $A_{ijt}^*$ 's as

$$A_{ijt}^* = \langle \boldsymbol{\beta}, \mathcal{X}_t \rangle [i, j] + \langle \boldsymbol{\theta}, \mathcal{G}_t \rangle [i, j] + s_{it} + r_{jt} + E_{ijt},$$

216 where  $s_{it}$ ,  $r_{jt}$ , and  $E_{ijt}$  are zero mean normal random variables (possibly correlated  
217 in complex ways, though letting  $s_{it}, r_{jt} \perp E_{ijt} \forall i, j, t$ ). Then we have the following  
218 proposition, the proof of which is given in Appendix 8.

219 **Proposition.**

$$\mathbb{P}(A_{ijt} = 1 | \boldsymbol{\beta}, \boldsymbol{\theta}) = \Phi \left( \frac{\mathbb{E}(A_{ijt}^*)}{\sqrt{\text{Var}(E_{ijt}) + \text{Var}(s_{it} + r_{jt})}} \right), \quad (3)$$

220 where  $\Phi(\cdot)$  is the CDF of a standard normal distribution, and  $\mathbb{E}(A_{ijt}^*)$  is given in  
221 Equation (1).

222 Now consider the very simple example where we have

$$\begin{pmatrix} s_{it} \\ r_{jt} \end{pmatrix} \stackrel{iid}{\sim} N \left( \mathbf{0}, \begin{pmatrix} \tau_s & 0 \\ 0 & \tau_r \end{pmatrix} \right)$$

223 and constant variance for the  $E_{ijt}$ 's. We can quickly see that should we  
224 ignore simultaneous dependence, any attempts to estimate  $(\boldsymbol{\beta}, \boldsymbol{\theta})$  would in fact  
225 unintentionally lead to the attenuated estimation of  $(\boldsymbol{\beta}, \boldsymbol{\theta})$  scaled by  $\text{Var}(E_{ijt}) + \tau_s + \tau_r$ .  
226 For more general cases, when  $\text{Var}(s_{it} + r_{jt})$  is time dependent or dependent on the  
227 actors  $i$  and  $j$ , it is unclear what, if anything, any naive estimates of  $(\boldsymbol{\beta}, \boldsymbol{\theta})$  are actually  
228 estimating.

#### 229 *2.4 Simultaneous and temporal autoregressive model*

230 A middle ground between fully ignoring simultaneous dependence and using a  
231 saturated covariance matrix  $\Sigma_{A^*, t}$  would be to assume that there ought to be some  
232 connection with the covariance between two dyads and the actors that are incident  
233 on those two dyads. This simple and intuitive idea will eventually lead us to a model  
234 resembling the social relations model (Warner *et al.*, 1979), having the form

$$A_{ijt}^* = \text{mean structure} + \text{sender effects} + \text{receiver effects} + \text{residuals}$$

235 (the final form is given in Equation (10)). To get there, we begin by introducing the  
236 following definition.

237 **Definition.** An  $n \times n$  matrix  $A^*$  has a *role-based additive covariance structure* if

$$\begin{aligned} \text{Cov}(A_{ij}^*, A_{k\ell}^*) &= \Sigma_s [i, k] + \Sigma_r [j, \ell] + \Sigma_{sr} [i, \ell] + \Sigma_{sr} [k, j] + \sigma_R^2 1_{\{(i,j)=(k,\ell)\} \cup \{(i,j)=(\ell,k)\}} \\ &+ \sigma_\epsilon^2 1_{\{(i,j)=(k,\ell)\}}, \end{aligned} \quad (4)$$

238 where  $\Sigma_s$ ,  $\Sigma_r$ , and  $\Sigma_{sr}$  are  $n \times n$  covariance matrices that represents, respectively, the  
239 covariance among the senders of the dyads, the receivers of the dyads, and between  
240 the senders and the receivers, and where  $\sigma_R^2$  and  $\sigma_\epsilon^2$  correspond to pair and dyad  
241 variance, respectively.

242 A role-based additive covariance structure can be interpreted to mean that the  
243 covariance between any two dyads  $(i, j)$  and  $(k, \ell)$  can be explained by how similar

244  $i$  and  $k$  are as senders, how similar  $j$  and  $\ell$  are as receivers, how  $i$  and  $\ell$  relate  
 245 to each other as sender and receiver, respectively, and similarly for  $k$  and  $j$ , the  
 246 variability due to reciprocated dyads, and the inherent variability between the  
 247 dyads.

248 The role-based additive covariance structure has a nice representation that  
 249 lends itself well to estimation. To demonstrate this, we provide the following  
 250 theorem.

251 **Theorem.** *The following are equivalent.*

252 1. The  $A_{ijt}^*$ 's are jointly normal with a role-based additive covariance structure and  
 253 mean given by Equation (1).

254 2.  $\mathcal{A}_t \sim N(\text{vec}(\langle \boldsymbol{\beta}, \mathcal{X}_t \rangle) + \langle \boldsymbol{\theta}, \mathcal{G}_t \rangle)$ ,

$$\begin{aligned}
 & J_n \otimes \Sigma_{st} + \Sigma_{rt} \otimes J_n + \mathbb{1}_n \otimes \Sigma_{srt} \otimes \mathbb{1}'_n + \mathbb{1}'_n \otimes \Sigma'_{srt} \otimes \mathbb{1}_n \\
 & + \sigma_R^2 M_R + (\sigma_\epsilon^2 + \sigma_R^2) I_{n^2},
 \end{aligned} \tag{5}$$

255 where  $\mathbb{1}_k$  is the  $k \times 1$  vector of 1's,  $J_k$  equals  $\mathbb{1}_k \mathbb{1}'_k$ , and  $I_k$  is the  $k \times k$  identity  
 256 matrix, and where  $M_R$  is a matrix such that for  $1 \leq i \neq j \leq n$ ,  $M_R[(j-1)n +$   
 257  $i, (i-1)n + j] = 1$  and  $M_r[\ell, m] = 0$  everywhere else.

258 3.  $A_t^* = \langle \boldsymbol{\beta}, \mathcal{X}_t \rangle + \langle \boldsymbol{\theta}, \mathcal{G}_t \rangle + \mathbf{s}_t \mathbb{1}' + \mathbb{1} \mathbf{r}' + E_t$ , where

$$\begin{aligned}
 & \begin{pmatrix} \mathbf{s}_t \\ \mathbf{r}_t \end{pmatrix} \stackrel{iid}{\sim} N \left( \mathbf{0}, \begin{pmatrix} \Sigma_{st} & \Sigma_{srt} \\ \Sigma'_{srt} & \Sigma_{rt} \end{pmatrix} \right), \\
 & (E_t[i, j], E_t[j, i])' \stackrel{iid}{\sim} N(\mathbf{0}, \sigma_\epsilon^2 I_2 + \sigma_R^2 J_2).
 \end{aligned} \tag{6}$$

259 The proof is given in Appendix 8.

260 Unconstrained, the covariance structure of Equation (6) still has  $O(n^2)$  parameters  
 261 to be estimated. The question then is how to appropriately, yet parsimoniously,  
 262 represent the covariance structure of  $(\mathbf{s}_t, \mathbf{r}_t)$ . In response, we pose the following  
 263 question: if the features found in  $(A_{t-1}, A_{t-2}, \dots)$  can appropriately capture the  
 264 temporal dependence through the mean structure, may we not also capitalize on  
 265 the information stored in  $(A_{t-1}, A_{t-2}, \dots)$  to estimate the simultaneous dependence  
 266 through the covariance structure? (This is similar in principle to ARCH models. See  
 267 Engle, 1982). We propose using an autoregressive model on the covariance structure  
 268 of  $(\mathbf{s}_t, \mathbf{r}_t)$  as well as on the mean structure of  $A_t^*$ , so that  $\text{Cov}(\mathcal{A}_t | \mathcal{A}_{t-1}, \mathcal{A}_{t-2}, \dots)$  is  
 269 some function of  $(\mathcal{A}_{t-1}, \mathcal{A}_{t-2}, \dots)$ .

270 Specifically, we consider  $\text{Cov}(\mathbf{s}_t, \mathbf{r}_t)$  with the following structure:

$$\Sigma_{st} = \sum_{k=1}^{K_s} \tau_{sk} H_{skt} \quad \Sigma_{rt} = \sum_{k=1}^{K_r} \tau_{rk} H_{rkt} \quad \Sigma_{srt} = \sum_{k=1}^{K_{sr}} \tau_{srk} H_{srkt} \tag{7}$$

272 where  $\tau_{sk}$ ,  $\tau_{rk}$ , and  $\tau_{srk}$  are positive valued parameters,  $H_{skt}$ ,  $H_{rkt}$ , and  $H_{srkt}$  are  
 273 functions of  $(A_{t-1}, A_{t-2}, \dots)$ , and  $H_{skt}, H_{rkt} \in \mathbf{S}_+^n$  for all  $k$ . Here,  $\mathbf{S}_+^n$  denotes the

274 positive semi-definite (PSD) cone. Writing  $\text{Cov}(s_t, r_t)$  in this manner, i.e., as a linear  
 275 combination of PSD matrices, is similar in principle to covariance structures studied  
 276 for many decades (e.g., Anderson, 1973). Constructing the covariance matrices in this  
 277 manner allows us to use the data to represent complex simultaneous dependence,  
 278 while reducing the number of parameters from  $O(n^2)$  to  $K_s + K_r + K_{sr}$ .

279 Note that this does not automatically ensure that  $\Sigma_{A^*,t} \in \mathbf{S}_+^{n^2}$ , and so some care  
 280 is still needed. To ensure that we have a valid covariance matrix, we constrain  
 281  $K_{sr} \leq \min\{K_s, K_r\}$ , and for  $1 \leq k \leq K_{sr}$  impose the constraint that

$$\begin{pmatrix} \tau_{sk} H_{skt} & \tau_{srk} H_{srkt} \\ \tau_{srk} H'_{srkt} & \tau_{rk} H_{rkt} \end{pmatrix} \in \mathbf{S}_+^{(2n)}. \quad (8)$$

282 The structure found in Equation (7) allows us to further decompose  $s_t$  and  $r_t$  as

$$\begin{aligned} 283 \quad s_t &= \sum_{k=1}^{K_s} s_{kt}, & s_{kt} &\stackrel{\text{ind}}{\sim} N(\mathbf{0}, \tau_{sk} H_{skt}) \\ r_t &= \sum_{k=1}^{K_r} r_{kt}, & r_{kt} &\stackrel{\text{ind}}{\sim} N(\mathbf{0}, \tau_{rk} H_{rkt}) \end{aligned} \quad \text{Cov}(s_{kt}, r_{k't}) = \begin{cases} \tau_{srk} H_{srkt} & \text{if } 1 \leq k = k' \leq K_{sr} \\ 0 & \text{otherwise.} \end{cases} \quad (9)$$

284 This then results in having our multivariate probit model with role-based additive  
 285 covariance structure represented as

$$A_t^* = \langle \boldsymbol{\beta}, \mathcal{X}_t \rangle + \langle \boldsymbol{\theta}, \mathcal{G}_t \rangle + \left( \sum_{k=1}^{K_s} s_{kt} \right) \mathbb{1}' + \mathbb{1} \left( \sum_{k=1}^{K_r} r_{kt} \right)' + E_t. \quad (10)$$

## 286 **2.5 Broader context of sender/receiver effects**

287 By first assuming an intuitive form for the covariance of the dyads, we are able to  
 288 arrive at a multivariate mixed effects probit model for the dynamic network, using  
 289 individual sender and receiver effects. The use of individual sender and receiver effects  
 290 has a long history in network analysis, starting with Warner *et al.* (1979). In nearly  
 291 all cases, the additive sender and receiver effects can be put within the framework  
 292 described above by setting  $K_s = K_r = K_{sr} = 1$  and  $H_{s1t} = H_{r1t} = H_{sr1} = I_n$ .  
 293 An important work using this is the  $p_2$  model of Duijn *et al.* (2004). This work  
 294 was built off of the  $p_1$  model of Holland & Leinhardt (1981), which was not  
 295 motivated by modeling an appropriate covariance structure. Latent space models  
 296 have incorporated additive sender/receiver effects as well, such as Hoff (2005) (which  
 297 also incorporated multiplicative effects), and Krivitsky *et al.* (2009).

298 The above-referenced works are all concerned with static networks.  
 299 Westveld & Hoff (2011) used the ideas of sender and receiver effects to model  
 300 the covariance of the data for dynamic networks. As with the others, they constrain  
 301  $K_s = K_r = K_{sr} = 1$  and  $H_{s1t} = H_{r1t} = H_{sr1} = I_n$ , while also assuming AR  
 302 processes on the sender and receiver effects (and on the residuals). While there  
 303 is merit in this approach, we still prefer capturing the temporal dependency  
 304 through the observation-driven model. This is primarily because one may utilize



305 specific network features, such as stability, reciprocity, transitivity, etc. to help  
 306 explain the temporal dependencies. In this way, one may argue that there is more  
 307 flexibility, and researchers can investigate the specific effects of various network  
 308 features.

309 The way in which we use sender and receiver effects here differs in two important  
 310 ways from previous uses. First, the constraints on the covariance matrix of the dyads  
 311 are relaxed to allow  $\Sigma_{A^*,t}$  to be dense, thus generalizing the way that researchers have  
 312 in the past used sender and receiver effects in their models. Second, we incorporate  
 313 past data to make the parameter space parsimonious. That is, a dense covariance  
 314 matrix with  $O(n^4)$  unknowns can, by leveraging past information, be estimated using  
 315  $K_s + K_r + K_{sr}$  parameters. For an example of how we may do this in practice, see  
 316 Section 2.6.

## 317 **2.6 An example of operationalization**

318 One of the strengths of Equations (1) and (7) is the flexibility in choosing the  
 319 features of the previous adjacency matrices to be used in constructing the mean  
 320 and covariance functions. In this subsection, we provide an example, based on  
 321 sociological principles as well as previous research in statistical models for networks,  
 322 with the intention that researchers using the STAR model may use whatever network  
 323 features are most appropriate for their particular context.

324 Fortunately for the analyst looking at dynamic network data, there has been much  
 325 focus in the social science literature on the salient structures of networks. To quote  
 326 Wasserman & Faust (1994),

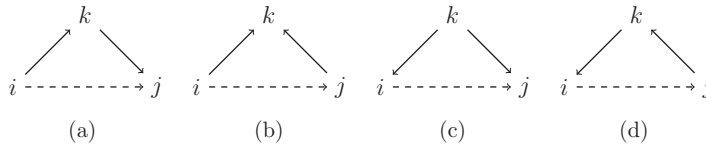
327 Many researchers have shown, using empirical studies, that social network data possess strong  
 328 deviations from randomness. ... data often fail to agree with predictions from [models with  
 329 assumptions, such as equal popularity, lack of transitivity, or no reciprocity].

330 Krackhardt & Handcock (2007) made note that it has long been argued that “the  
 331 triad, not the dyad, is the fundamental social unit that needs to be studied” (see  
 332 also Simmel & Wolff, 1950), which further emphasizes that transitivity is, to quote  
 333 Wasserman & Faust (1994) again, “indeed a compelling force in the organization of  
 334 social groups.”

335 These notions then motivate the construction of  $\mathcal{G}_t$ , the three-dimensional tensor  
 336 whose  $\ell$ th slice is denoted by  $\mathcal{G}_{\ell t}$ , as given in Table 1. We can categorize these  
 337 eight structures of the network in the following terms.  $\mathcal{G}_{1t}$  and  $\mathcal{G}_{2t}$  correspond to  
 338 first-order structures, that is, features of the network that relate to individual actors  
 339 only.  $\mathcal{G}_{3t}$  and  $\mathcal{G}_{4t}$  correspond to second-order structures, that is, features of the  
 340 network that relate to dyads.  $\mathcal{G}_{5t}$  to  $\mathcal{G}_{8t}$  correspond to third-order structures, that is,  
 341 features of the network that relate to triads. In particular,  $\mathcal{G}_{5t}$  to  $\mathcal{G}_{7t}$  correspond to  
 342 transitivity in the network, i.e., the probability that a transitive relation exists, while  
 343  $\mathcal{G}_{8t}$  corresponds to a cycle, i.e., the probability that a three-cycle will be completed.  
 344 These last four structures are depicted visually in Figure 1, where we are considering  
 345 the probability of an edge from  $i$  to  $j$  and visualizing the transitive and cyclic triadic  
 346 relations involving the third actor  $k$ . One note regarding  $\mathcal{G}_{1t}$  to  $\mathcal{G}_{8t}$  is that these  
 347 same features could of course be trivially extended to more than just a lag of 1  
 348 whenever appropriate.

Table 1. Example of how to construct  $\mathcal{G}_t$ , incorporating first-, second-, and third-order structures.

(out degree)	$\mathcal{G}_{1t} = A_{t-1}J_n$	$\mathcal{G}_{1t}[i, j] = \sum_{k=1}^n A_{ik(t-1)}$
(in degree)	$\mathcal{G}_{2t} = J_n A_{t-1}$	$\mathcal{G}_{2t}[i, j] = \sum_{k=1}^n A_{kj(t-1)}$
(stability)	$\mathcal{G}_{3t} = A_{t-1}$	$\mathcal{G}_{3t}[i, j] = A_{ij(t-1)}$
(reciprocity)	$\mathcal{G}_{4t} = A'_{t-1}$	$\mathcal{G}_{4t}[i, j] = A_{ji(t-1)}$
(transitivity 1)	$\mathcal{G}_{5t} = A_{t-1}A_{t-1}$	$\mathcal{G}_{5t}[i, j] = \sum_{k=1}^n A_{ik(t-1)}A_{kj(t-1)}$
(transitivity 2)	$\mathcal{G}_{6t} = A_{t-1}A'_{t-1}$	$\mathcal{G}_{6t}[i, j] = \sum_{k=1}^n A_{ik(t-1)}A_{jk(t-1)}$
(transitivity 3)	$\mathcal{G}_{7t} = A'_{t-1}A_{t-1}$	$\mathcal{G}_{7t}[i, j] = \sum_{k=1}^n A_{ki(t-1)}A_{kj(t-1)}$
(cycle)	$\mathcal{G}_{8t} = A'_{t-1}A'_{t-1}$	$\mathcal{G}_{8t}[i, j] = \sum_{k=1}^n A_{ki(t-1)}A_{jk(t-1)}$

Fig. 1. Network structures which are being summed over  $k$  to determine the mean of  $A_{ijt}^*$ . (a)  $\mathcal{G}_{5t}[i, j]$ . (b)  $\mathcal{G}_{6t}[i, j]$ . (c)  $\mathcal{G}_{7t}[i, j]$ . (d)  $\mathcal{G}_{8t}[i, j]$ .

349 Intuitively,  $\Sigma_{st}$  and  $\Sigma_{rt}$  ought to reflect how similar actors behave as senders and  
 350 receivers, respectively. We therefore suggest setting  $K_s = K_r = 2$ ,  $K_{sr} = 1$ , and

$$\begin{aligned}
 H_{s1t} &= H_{r1t} = H_{sr1t} = I_n \\
 H_{s2t} &= D_{out,(t-1)}^{-1/2} A_{t-1} A'_{t-1} D_{out,(t-1)}^{-1/2}, \\
 H_{r2t} &= D_{in,(t-1)}^{-1/2} A'_{t-1} A_{t-1} D_{in,(t-1)}^{-1/2},
 \end{aligned} \tag{11}$$

351 where  $D_{out,(t-1)}$  and  $D_{in,(t-1)}$  are diagonal matrices whose diagonal entries are the  
 352 out-degrees and in-degrees of  $A_{t-1}$ , respectively. The  $(i, j)$ -th entry of  $H_{s2t}$  then is the  
 353 number of actors to whom both  $i$  and  $j$  sent edges scaled by the geometric mean of  
 354 the total number of actors to whom  $i$  and  $j$  each sent edges. In this manner, we are  
 355 capturing the intended notion of similarity between senders while enforcing  $H_{s2t}$   
 356 to be PSD. In fact,  $H_{s2t}$  is a valid correlation matrix. Similarly for  $H_{r2t}$ , a note on the  
 357 practical implementation of this is that to avoid the possibility of dividing by zero  
 358 anywhere, in our analyses we set the diagonal of  $A_{t-1}$  to be  $\mathbb{1}$  when computing  $H_{s2t}$   
 359 and  $H_{r2t}$ . To ensure that the covariance of  $(s_t, r_t)$  is PSD, and hence the covariance  
 360 of  $\mathcal{A}_t$  is PSD, we constrain

$$\Omega := \begin{pmatrix} \tau_{s1} & \tau_{sr1} \\ \tau_{sr1} & \tau_{r1} \end{pmatrix} \in \mathbf{S}_+^2. \tag{12}$$

## 361 2.7 Undirected networks

362 The above proposed methodology has focused on directed dynamic networks.  
 363 Simplifying to an undirected dynamic network implies that Equations (4) and (5)  
 364 can be written

$$\begin{aligned}
 \text{Cov}(A_{ijt}^*, A_{k\ell t}^*) &= \Sigma_{st}[i, k] + \Sigma_{st}[j, \ell] + \Sigma_{st}[i, \ell] + \Sigma_{st}[k, j] + \sigma^2 \mathbb{1}_{[(i,j)=(k,\ell)]} \\
 \Leftrightarrow \text{Cov}(\mathcal{A}_t) &= J_n \otimes \Sigma_{st} + \Sigma_{st} \otimes J_n + \mathbb{1} \otimes \Sigma_{st} \otimes \mathbb{1}' + \mathbb{1}' \otimes \Sigma_{st} \otimes \mathbb{1} + \sigma^2 I.
 \end{aligned} \tag{13}$$

The estimation algorithm given in Section 3 can be adapted to the undirected case; some of the details which are not obvious are given in Appendix 8. In the analysis of Section 7.1, we set

$$\Sigma_{st} = \tau_s H_{st}, \text{ where } H_{st} = D_{(t-1)}^{-1/2} A_{t-1} A_{t-1} D_{(t-1)}^{-1/2} \quad (14)$$

and  $D_t$  is the diagonal matrix whose diagonal entries are the degrees of the actors corresponding to  $A_t$ , i.e.,  $A_t \mathbf{1}$ . For autoregressive mean terms, we used

$$\begin{aligned} \text{(degree)} \quad \mathcal{G}_{1t} &= A_{t-1} J_n + J_n A_{t-1} \quad \mathcal{G}_{1t}[i, j] = \sum_{k=1}^n (A_{ik(t-1)} + A_{jk(t-1)}) \\ \text{(stability)} \quad \mathcal{G}_{2t} &= A_{t-1} \quad \mathcal{G}_{2t}[i, j] = A_{ij(t-1)} \\ \text{(triangle)} \quad \mathcal{G}_{3t} &= A_{t-1} A_{t-1} \quad \mathcal{G}_{3t}[i, j] = \sum_{k=1}^n A_{ik(t-1)} A_{jk(t-1)}. \end{aligned}$$

### 3 Variational Bayes estimation

From a Bayesian perspective, we would like to make posterior inference regarding the mean parameters  $\beta$  and  $\theta$  as well as the variance components  $\tau_{sk}$ 's,  $\tau_{rk}$ 's, and  $\tau_{srk}$ 's. In what follows, we will assume the particular formulation given in Section 2.6. Thus of interest is deriving  $\pi(\beta, \theta, \Omega, \tau_{s2}, \tau_{r2}, \sigma_R^2 | \{A_t\}_{t=0}^T)$ . Note that just as with any probit model,  $\sigma_\epsilon^2$  is constrained to equal 1 for identifiability. We assign the following priors on the model parameters.

$$\begin{aligned} (\beta', \theta')' &\sim N(\mathbf{0}, \text{diag}(\sigma_\beta^2, \dots, \sigma_\beta^2, \sigma_\theta^2, \dots, \sigma_\theta^2)), \\ \tau_{s2} &\sim IG(a_{s0}, b_{s0}), \\ \tau_{r2} &\sim IG(a_{r0}, b_{r0}), \\ \Omega &\sim IW(a_{\Omega 0}, B_{\Omega 0}), \\ \sigma_R^2 &\sim IG(a_{R0}, b_{R0}), \end{aligned}$$

where  $\text{diag}(\sigma_\beta^2, \dots, \sigma_\beta^2, \sigma_\theta^2, \dots, \sigma_\theta^2)$  is the  $(p_1 + p_2) \times (p_1 + p_2)$  diagonal matrix whose first  $p_1$  diagonal entries are  $\sigma_\beta^2$  and whose last  $p_2$  diagonal entries are  $\sigma_\theta^2$ ,  $IG(a, b)$  is the inverse gamma distribution with shape parameter  $a$  and scale parameter  $b$ , and  $IW(a, B)$  denotes the inverse Wishart distribution with degrees of freedom  $a$  and scale matrix  $B$ .

Rather than implementing a computationally expensive MCMC algorithm, we implement a mean field variational Bayes (VB) algorithm. This estimation technique finds an approximation of the posterior distribution such that the Kullback–Leibler divergence between this approximation and the true posterior distribution is minimized. This minimization is done under the constraint that the approximated posterior density is a product of densities corresponding to a partition of the unknown model parameters. See, e.g., Gelman *et al.* (2004) (Chapter 13) for a brief overview of variational methods.

While much faster than MCMC, one issue with the VB algorithm is a negative bias of the variance components. In our analyses, we found that the bias was so strong in  $\sigma_R^2$  as to render the reciprocity effects negligible, which led to poorer performance overall. To address this, first consider further data augmentation via the  $n \times n$  symmetric matrices of dyad-pair specific random effects  $R_t$ , such that  $R_t[i, j] = R_t[j, i] \stackrel{iid}{\sim} N(0, \sigma_R^2)$ . That is, we now have the equivalent form of

397 Equation (10)

$$A_t^* = \langle \boldsymbol{\beta}, \mathcal{X}_t \rangle + \langle \boldsymbol{\theta}, \mathcal{G}_t \rangle + \left( \sum_{k=1}^{K_s} s_{kt} \right) \mathbb{1}' + \mathbb{1} \left( \sum_{k=1}^{K_r} r_{kt} \right)' + R_t + \tilde{E}_t, \quad (15)$$

398 where  $\tilde{E}_t$  is a matrix of *iid* normal random variables with zero mean and variance  
 399  $\sigma_\epsilon^2$ . To prohibit  $\sigma_R^2$  from shrinking to zero, we treat it as a hyperparameter for the  
 400  $R_t$ 's. While not ideal, this seemed to improve overall performance.

401 The specific form of the approximated posterior is

$$\begin{aligned} & \pi(\boldsymbol{\beta}, \boldsymbol{\theta}, \tau_{s2}, \tau_{r2}, \boldsymbol{\Omega}, \sigma_R^2, \{A_t^*\}_{t=1}^T, \{s_{1t}, r_{1t}, s_{2t}, r_{2t}\}_{t=1}^T, \{R_t\}_{t=1}^T | \{A_t\}_{t=0}^T) \\ & \approx q_1(\boldsymbol{\beta}, \boldsymbol{\theta}) q_2(\tau_{s2}, \tau_{r2}, \boldsymbol{\Omega}) q_3(\{\mathcal{A}_t\}_{t=1}^T) q_4(\{s_{1t}, r_{1t}, s_{2t}, r_{2t}\}_{t=1}^T) q_5(\{R_t\}_{t=1}^T) q_6(\sigma_R^2). \end{aligned} \quad (16)$$

402 This is an iterative scheme, in which we use the parameters from, say,  $q_\ell$  to estimate  
 403  $q_m$  and vice versa. The closed-form solutions to the VB updates are given in  
 404 Appendix 8. The derivations for the sender and receiver effects are also provided, as  
 405 these are not straightforward due to the fact that the derivations must be taken with  
 406 respect to the distribution of  $A_t^* \circ (J_n - I_n)$  rather than  $A_t^*$ , as given in Equation (10).

407 The VB approach is quite fast and yields good point estimates. This comes at a  
 408 cost, however. VB algorithms may get stuck in local modes, and which local mode  
 409 one ends up in may be highly dependent on the starting values (see, e.g., Bickel *et al.*,  
 410 2013; Salter-Townshend & Murphy, 2013, for more detailed studies using variational  
 411 approaches). Additionally, by partitioning the parameters and forcing them to be  
 412 independent in the approximate posterior, the posterior probability regions are  
 413 typically much too concentrated. In our context, we found that a Gibbs sampler  
 414 obtained similar posterior means, though wider credible intervals. The MCMC  
 415 algorithm was simply too slow in practice for networks of medium to large size,  
 416 however.

#### 417 **4 Generalizing to weighted networks**

418 In this section, we demonstrate how to generalize our approach to weighted networks  
 419 in which the dyads are not constrained to  $\{0, 1\}$ . We accomplish this by placing our  
 420 work within the framework of a GLMM. Most researchers, statisticians or not, are  
 421 familiar with GLMMs that are often the tool of choice for modeling dependent non-  
 422 Gaussian data. The general framework assumes that a function of the means of the  
 423 random variables are themselves correlated (typically Gaussian) random variables,  
 424 thus allowing researchers to control for the correlation among the data. Specifically,  
 425 for some response vector  $\mathbf{y}$ , covariate matrix  $X$ , random variables  $\boldsymbol{\gamma}$ , and design  
 426 matrix  $Z$ , we write

$$g(\mathbb{E}(\mathbf{y})) = X\boldsymbol{\beta} + Z\boldsymbol{\gamma}. \quad (17)$$

427 (Note that the notation in Equation (17) is not linked to anything previously given,  
 428 but is rather a general form for a GLMM).

429 Up to this point, we have assumed a probit model, as this was a natural approach  
 430 to dealing with complex dependencies in binary data. This is equivalent to a GLMM  
 431 using the normal inverse cumulative distribution function as the link function  $g$ .

432 Placing our proposed methods within the GLMM framework allows us to use other  
 433 link functions such as a  $\text{logit}()$  for logistic regression, as well as allowing us to  
 434 model other types of non-Gaussian data; e.g., should our network data be count,  
 435 as is often the case, we may use a log link corresponding to a Poisson or Negative  
 436 Binomial family of distributions. Countless texts describe these models, and in fact  
 437 GLMMs are so prevalent that many fields have books or articles demonstrating  
 438 how to apply GLMMs to their specific subject area (e.g., Bolker *et al.*, 2009; Gbur,  
 439 2012; Krueger & Montgomery, 2014; Bharadwaj, 2016).

440 We wish to maintain the covariance structures detailed in Section 2.4, and in  
 441 particular that implied by Equation (15) but generalize it to other link functions  
 442 and other data types. This can be done by setting

$$\begin{aligned}
 & g\left(\mathbb{E}(\mathcal{A}_t | \mathcal{A}_{t-1}, \mathcal{A}_{t-2}, \dots)\right) \\
 &= (\text{vec}^-(X_{1t}), \text{vec}^-(X_{2t}), \dots, \text{vec}^-(\mathcal{G}_{1t}), \text{vec}^-(\mathcal{G}_{2t}), \dots) \begin{pmatrix} \beta \\ \theta \end{pmatrix} + Z\gamma_t, \\
 Z &= (\mathbb{1}'_{k_s} \otimes Z_s \quad \mathbb{1}'_{k_r} \otimes Z_r \quad Z_{rec}), \\
 \gamma_t &= (s'_{1t} \quad \dots \quad s'_{k_{st}} \quad r'_{1t} \quad \dots \quad r'_{k_{rt}} \quad \mathcal{R}'_t)', \tag{18}
 \end{aligned}$$

443 where  $\mathcal{R}_t$  contains the lower triangular elements of  $R_t$  (i.e.,  $\mathcal{R}_t = (R_{21t}, R_{31t}, \dots,$   
 444  $R_{n(n-1)t})$ ), and where  $\text{vec}^-(M)$  for some  $n \times n$  square matrix  $M$  is the standard  
 445  $\text{vec}(M)$ , while omitting the diagonals; hence  $\text{vec}^-(M)$  will be an  $n(n-1) \times 1$  vector.  
 446 To construct  $Z_s$ , we may stack  $I_{n,(-1,\cdot)}$ ,  $I_{n,(-2,\cdot)}$ ,  $\dots$ , and  $I_{n,(-n,\cdot)}$  to form a  $n(n-1) \times n$   
 447 matrix, where  $I_{n,(-i,\cdot)}$  is the  $n \times n$  identity matrix with the  $i$ th row removed.  $Z_r$  is  
 448 simply  $I_n \otimes \mathbb{1}_{n-1}$ . Constructing the  $n(n-1) \times n(n-1)/2$  matrix  $Z_{rec}$  is perhaps the  
 449 most involved, but can be accomplished by the following pseudocode:

Set all elements of  $Z_{rec}$  to 0.

```

for  $i \in \{1, 2, \dots, n\}$  do
    for  $j \in \{1, 2, \dots, n\} \setminus i$  do
         $r \leftarrow (n-1)(j-1) + i - 1_{[i>j]}$ 
        if  $i > j$  then  $c = n(j-1) - \frac{j(j+1)}{2} + i$  else  $c = n(i-1) - \frac{i(i+1)}{2} + j$ 
         $Z_{rec}[r, c] \leftarrow 1$ 
    end
end
    
```

451 By placing our methods within the GLMM framework, we provide an easy way  
 452 to handle a wide range of data types as well as overdispersion.

### 5 Evidence of simultaneous dependence

453 We now begin to address determining whether or not simultaneous dependence  
 454 exists. Just as with mixed models, we could check the intraclass correlation between  
 455 the pairs of residuals  $E_t[i, j]$  and  $E_t[j, i]$  to evaluate the importance of simultaneous  
 456 reciprocity. That is, estimate

$$\frac{\sigma_R^2}{\sigma_R^2 + 1}. \tag{19}$$

458 The issue is not so straightforward for the other types of simultaneous dependence.  
 459 Consider the case where the variance of  $A_{ijt}^*$  does not depend on the actors  $i$  and

460  $j$  nor the time  $t$ , the off diagonals of  $H_{srk}$  are 0 for all  $k$ , and the  $H_{sk}$ 's and  $H_{rk}$ 's  
 461 have been scaled such that the diagonal entries are 1 (as is true in our example of  
 462 Section 2.6). Then analogously to Equation (19), one may consider the vector

$$\mathbf{v}/(\mathbf{v}'\mathbf{1}) \quad \text{where} \quad \mathbf{v} = (\tau_{s1}, \tau_{s2}, \dots, \tau_{sK_s}, \tau_{r1}, \dots, \tau_{rK_r}, \sigma_R^2, 1). \quad (20)$$

463 Though Equation (20) appears similar to a vector of intraclass correlations,  
 464 these two things are in fact not comparable. Equation (20) is only a  
 465 ratio of variance components, while Equation (19) is a veritable correlation.  
 466 In the context of a directed network, there are seven correlations  
 467 we could consider:  $Cor(A_{ijt}^*, A_{k\ell t}^*)$ ,  $Cor(A_{ijt}^*, A_{kit}^*)$ ,  $Cor(A_{ijt}^*, A_{kjt}^*)$ ,  $Cor(A_{ijt}^*, A_{i\ell t}^*)$ ,  
 468  $Cor(A_{ijt}^*, A_{ij\ell}^*)$ ,  $Cor(A_{ijt}^*, A_{j\ell t}^*)$ , and  $Cor(A_{ijt}^*, A_{jit}^*)$ . Moreover, these seven correlations  
 469 very well may differ based on which actors we are considering! Instead, we present a  
 470 visualization method that may be used to assess the evidence regarding the existence  
 471 and impact of simultaneous dependence.

472 The main idea is that we would like to evaluate how much of our posterior  
 473 distributions of  $(\{s_{kt}\}_{k=1}^{K_s}, \{r_{kt}\}_{k=1}^{K_r})$ ,  $t = 1, \dots, T$ , are located within some small ball  
 474 around zero. If there is no simultaneous dependence, then we would expect the  
 475 posterior distributions to reflect this in having most of their mass near zero. Hence,  
 476 we are concerned with

$$\begin{aligned} \mathcal{P}_{\epsilon,t} &:= \int_{\mathcal{B}_\epsilon} dF \left( \{s_{kt}\}_{k=1}^{K_s}, \{r_{kt}\}_{k=1}^{K_r} \mid \{A_t\}_{t=1}^T \right) \\ &= \mathbb{P} \left( \|(s'_{1t}, \dots, s'_{K_s t}, r'_{1t}, \dots, r'_{K_r t})\| < \epsilon \mid \{A_t\}_{t=1}^T \right), \end{aligned} \quad (21)$$

477 where  $\mathcal{B}_\epsilon$  represents the ball around zero of radius  $\epsilon$ . This probability is very easily  
 478 and accurately estimated using a Monte Carlo approximation using draws from  
 479  $q_4$ . We can then plot  $\mathcal{P}_{\epsilon,t}$  vs.  $\epsilon$  to obtain a visualization of the magnitude of our  
 480 individual effects at each time point.

481 Our estimate of this high-dimensional posterior distribution,  $q_4$ , has the surprising  
 482 characteristic that most of the probability mass lies within a thin shell far from the  
 483 posterior mean (intuitively, this is because the volume of  $\mathcal{B}_\epsilon$  grows exponentially  
 484 with  $n$ ). Therefore, we need some comparison for the  $\mathcal{P}_{\epsilon,t}$ 's. It may be helpful to  
 485 compare the posterior for  $\|(s'_{1t}, \dots, s'_{K_s t}, r'_{1t}, \dots, r'_{K_r t})\|$  with the distribution of the  
 486 magnitude of a  $N(\mathbf{0}, \frac{p(\sigma_R^2+1)}{(1-p)(K_s+K_r)} J_{n(K_s+K_r)})$  random variable for some  $p \in (0, 1)$ . The  
 487 distribution of this comparative random variable arises from letting the ratio of  
 488 variances in Equation (20) sum to a proportion  $p$  for these simultaneous dependence  
 489 terms (and letting each of the  $K_s + K_r$  terms contribute equally); that is, what does  
 490 the distribution of  $\|(s'_{1t}, \dots, s'_{K_s t}, r'_{1t}, \dots, r'_{K_r t})\|$  look like if simultaneous dependence  
 491 accounts for  $p(100)\%$  of the variance of the  $A_{ijt}^*$ 's compared with the inherent noise?  
 492 Though there well may be better comparative distributions, what we have described  
 493 provides a reasonable frame of reference by which we may evaluate the strength  
 494 of the evidence of simultaneous dependence as given by the posterior distribution  
 495 for the sender and receiver effects. By looking at the visualization rather than just  
 496 the ratio of variance components, we do not throw away the effects of the off-  
 497 diagonal elements of the covariance matrices  $\Sigma_{st}$  and  $\Sigma_{rt}$  nor the entirety of  $\Sigma_{srt}$   
 498 when evaluating the evidence of the existence of simultaneous dependence.

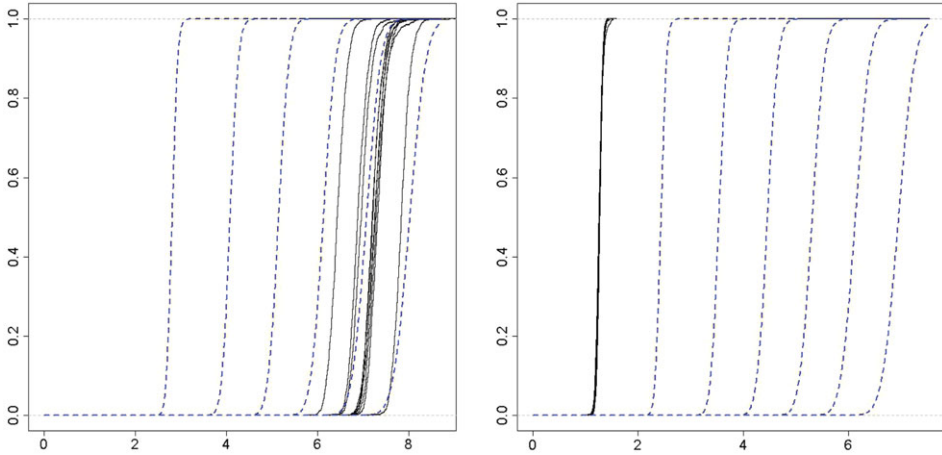


Fig. 2. Empirical example of the visualization of the existence of simultaneous dependence. The horizontal axis corresponds to the  $\epsilon$  radius of a ball  $\mathcal{B}_\epsilon$  about zero, and the vertical axis is  $\mathcal{P}_{\epsilon,\cdot}$ . Each solid line corresponds to a time point ( $T = 10$ ), and the dotted lines correspond to the comparative random variable having proportion of variance attributable to simultaneous dependence of, from left to right,  $p = 0.05, 0.1, 0.15, 0.2, 0.25, 0.3$ . The left panel corresponds to data generated with simultaneous dependence and the right panel without.

The distribution of the magnitude of the comparative random variable can be evaluated in the following way. Let  $\mathbf{x} \sim N_n(\mathbf{0}, \sigma^2 I_n)$  (e.g.,  $\sigma^2 = p(\sigma_R^2 + 1)/((1-p)(K_s + K_r))$ ). Then, let  $Y^2 := \mathbf{x}'\mathbf{x}/\sigma^2 \sim \chi^2(n)$ . Then  $Y \sim \chi(n)$  and thus

$$\mathbb{P}(\|\mathbf{x}\| \leq \epsilon) = \mathbb{P}(Y \leq \frac{\epsilon}{\sigma}) = \frac{\gamma(n/2, (\epsilon/\sigma)^2/2)}{\Gamma(n/2)}, \tag{22}$$

where  $\gamma(\cdot, \cdot)$  is the lower incomplete gamma function. Using this we can directly compute  $\mathcal{P}_\epsilon$  corresponding to this comparative random variable.

Figure 2 provides an empirical demonstration of the proposed visualization technique using the results from an arbitrarily chosen simulated data set as described in Section 6; note that we used the variance of the estimated  $R_t$ 's as a proxy for  $\sigma_R^2$ . The left panel corresponds to data generated with simultaneous dependence and the right panel without. The solid lines correspond to the individual effects at a particular time point, and the dotted lines correspond to the comparative noise for  $p \in \{0.05, 0.1, 0.15, 0.2, 0.25, 0.3\}$ .

## 6 Simulation study

We performed a simulation study in order to investigate two things. First, what is the effect of ignoring simultaneous dependence when it exists? Second, what is the effect of modeling simultaneous dependence when it does not exist? Specifically, we wish to investigate the effects on the mean parameters, as these will typically be the parameters of interest to the researcher. To this end, we simulated 100 network data sets where there was simultaneous dependence and 100 without such dependencies.

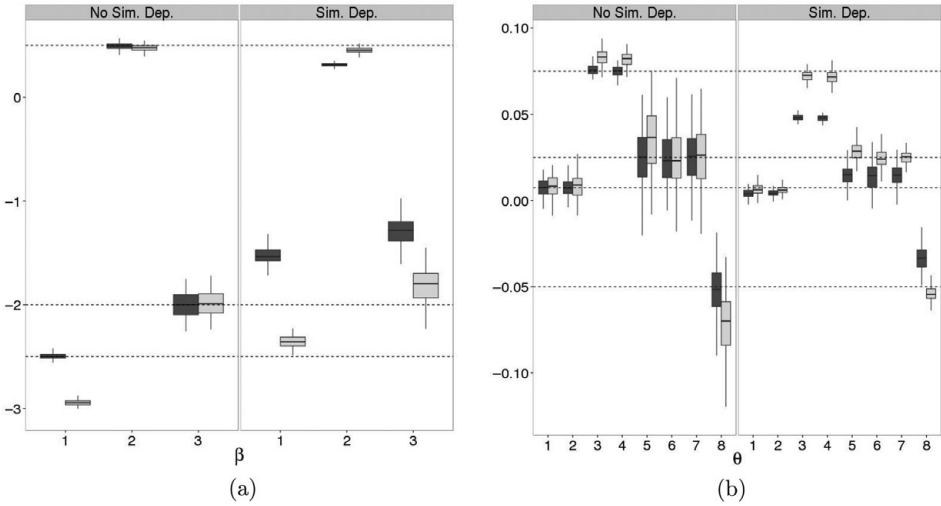


Fig. 3. Posterior means of (a)  $\beta$  and (b)  $\theta$  from analyzing the simulated data sets described in Section 6. Note that  $\theta_3$  and  $\theta_4$  have been scaled by  $1/10$  for visualization purposes. Horizontal dotted lines indicate true values of the parameters; the true  $\beta$  equals  $(-2.5, 0.5, -2)$ , and the true  $\theta$  equals  $(0.0075, 0.0075, 0.75, 0.75, 0.025, 0.025, 0.025, -0.05)$ . Lightly shaded boxplots correspond to accounting for simultaneous dependence in the model; dark shaded boxplots correspond to ignoring the simultaneous dependence.

518 For each of these 200 data sets, we fit two models, one accounting for and the other  
 519 ignoring these dependencies.

520 Each simulated data set had  $n = 100$  and  $T = 10$ . We incorporated two covariates  
 521 as well as an intercept (i.e.,  $p_1 = 3$ ). The first dyadic covariate was a binary  
 522 variable taking values 0 or 1 with equal probability; this covariate was treated as  
 523 constant over time. The second covariate was constructed by first simulating  $n$  AR(1)  
 524 processes with autoregressive coefficient equal to 0.9 and transition variance equal  
 525 to 0.05, and then at each time point taking the distance between the corresponding  
 526 cross-sectional views of the AR(1) time series. The coefficients were then set to be  
 527  $\beta = (-2.5, 0.5, -2)$  for the intercept, first covariate, and second covariate, respectively.  
 528 We set  $\theta = (0.0075, 0.0075, 0.75, 0.75, 0.025, 0.025, 0.025, -0.05)$ , corresponding to  
 529  $\mathcal{G}_{1t}, \dots, \mathcal{G}_{8t}$ , respectively, where the  $\mathcal{G}_{it}$ 's are as given in Section 2.6. Note that  $\theta_3$  and  
 530  $\theta_4$  needed to be on different scales, as these were the only coefficients corresponding  
 531 to network structures taking values in  $\{0, 1\}$  rather than  $\{0, 1, \dots, n-1\}$ . For the  
 532 simulations with simultaneous dependence, we set  $\tau_{s2} = 0.2$ ,  $\tau_{r2} = 0.1$ , the diagonal  
 533 of  $\Omega$  to be  $(0.25, 0.5)$ , the off-diagonals of  $\Omega$  equal to 0.1, and  $\sigma_R^2 = 0.5$ .

534 The results are given graphically in Figure 3. Figure 3(a) shows the boxplots of the  
 535 estimates of the  $3 \times 1$  vector  $\beta$ . The columns correspond to the true model, and the  
 536 shade of the boxplots corresponds to whether or not simultaneous dependence was  
 537 accounted for. From this, we see that in the presence of simultaneous dependence,  
 538 our proposed approach does a much better job at estimating the true values of  $\beta$   
 539 than when the simultaneous dependence is ignored. In the absence of simultaneous  
 540 dependence, with the exception of the intercept (arguably of little importance in



most research settings) our proposed approach performs very comparably to the models that ignore simultaneous dependence. We can reach the same conclusions looking at Figure 3(b), which gives the boxplots of the estimates of the  $8 \times 1$  vector  $\theta$ .

In summary, accounting for simultaneous dependence in the model is extremely important in obtaining more accurate estimates of the coefficients in the mean function, and doing so even in the absence of simultaneous dependence does not seem to do much harm in the estimation. If concerns persist, one may perform the visualization described previously, as seen in Figure 2, to determine whether or not to include simultaneous dependence in the final model.

## 7 Data analyses

We now look at two real data sets with the intent of illustrating how our approach can be implemented in practice both for directed and undirected data. In the last example, we illustrate the change in impact from simultaneous dependence as the time intervals vary from fine to coarse.

### 7.1 Conference proximity network

We first look at a proximity network taken from conference goers at The Last Hope Conference, collected and made available by the OpenAMD Project (OpenAMD, 2008). The 2008 conference goers had the option to wear an RFID badge, which tracked their movements throughout the conference. Thus, we are able to construct a proximity network, connecting two actors if they spent time close to one another. This type of network is quite important in, e.g., infectious disease (Vanhems *et al.*, 2013) and the study of human behavior and organization (Eagle & Pentland, 2006). Our undirected network data consisted of 1,190 actors over 29 hours (i.e.,  $T = 29$ ). We set  $A_{ijt}(= A_{jit})$  to be 1 if actors  $i$  and  $j$  visited the same location during the  $t$ th hour.

Figure 4(a) shows the evidence of simultaneous dependence. From this plot, we see that there is very strong evidence of such dependencies even though the time intervals are rather fine (1 hour). Figure 4(b) shows the posterior means for the autoregressive terms when ignoring simultaneous dependence (dark gray) and when accounting for it (light gray). Notice that the estimates are, with the exception of stability, quite different; indeed, ignoring simultaneous dependence leads to a negative estimate for the effect of triangles, which seems very unlikely given previous work done on structural balance theory.

### 7.2 World trade data

The second data set that we consider here is that of a world trade network. We let  $A_{ijt}$  be 1 if country  $i$  exports to country  $j$  at time  $t$ . This data were collected from the Correlates of War Project (Barbieri & Keshk, 2012; Barbieri *et al.*, 2009). Along with the export/import data, we used as covariates religious makeup of a country (Maoz & Henderson, 2013), defense pacts, neutrality pacts, non-aggression pacts, and ententes (Gibler, 2009). We analyze this data in two ways. First, we focus on a larger number of countries that exist over recent years. We then look at a smaller subset of countries that all exist over a longer period of time and look at how the evidence for simultaneous dependence changes as the time intervals get coarser.

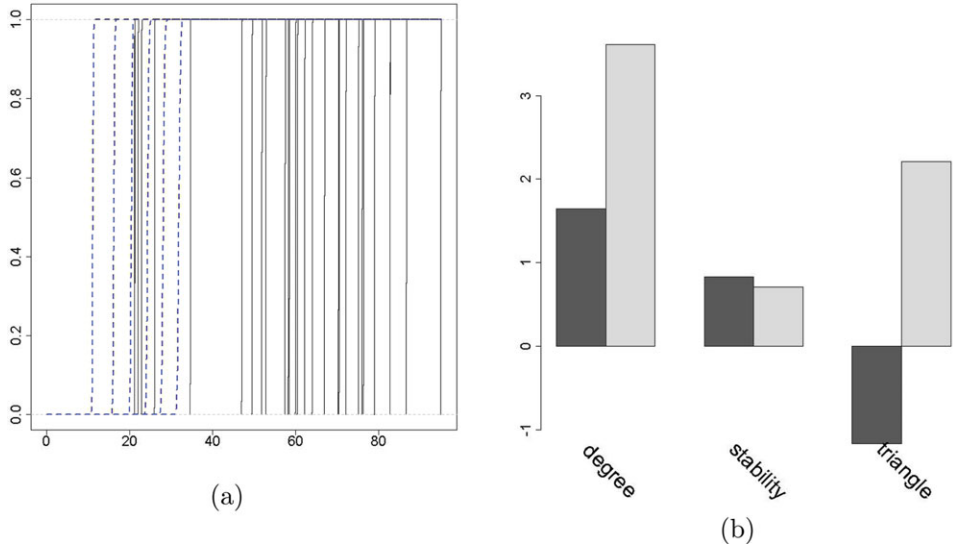


Fig. 4. Results from the AMD proximity network data. (a) Plot of  $\mathcal{P}_{\epsilon,t}$  vs.  $\epsilon$ . Each solid curve corresponds to the individual effects from a particular time point. The dotted lines correspond to the comparative random variable setting  $p = 0.05, 0.1, 0.15, 0.2, 0.25, 0.3$ . See Section 5 for details. (b) Posterior means for the coefficients of  $\theta$ . Dark gray indicates ignoring simultaneous dependence, while light gray indicates accounting for this dependence in the model.

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7.2.1 179 nations from 1993 to 2009

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We consider all countries that exist and are involved in trade on an annual basis over the period from 1993 to 2009. For each of these countries, we have the measurements of the proportion of their population that belongs to each of the main world religions and the sub-branches of these religions (a total of 30 categories). These measurements only occur once every 5 years, which we interpolated to construct annual religious data. We then constructed the dyadic covariates by taking the Hellinger distance of two multinomial distributions whose probability vectors equal those nations' vector of proportions of religious adherents. Letting  $p_{it}$  be the  $30 \times 1$  vector of the  $i$ th nation's proportion of religious adherents, this is equivalent to setting the dyadic covariate between  $i$  and  $j$  equal to  $\sqrt{1 - \sum_{r=1}^{30} \sqrt{p_{itr}p_{jtr}}}$ . The four types of pacts each were simply binary variables indicating whether or not countries  $i$  and  $j$  were engaged in such a pact during year  $t$ .

Figure 5(a) depicts the evidence of simultaneous dependence. From this, we see that we there is evidence of non-negligible simultaneous dependence, though much less so than in the AMD network data. Figure 5(b) shows the posterior means for the covariates and Figure 5(c) shows the same for the autoregressive terms, where again dark gray indicates ignoring simultaneous dependence and light gray indicates accounting for it in the model. As is consistent with the simulation results, when there is weaker simultaneous dependence in the data, these estimates are more in agreement. There are still some differences, mostly manifested in the attenuation of the estimates as well as more dramatic differences in the triadic effects.

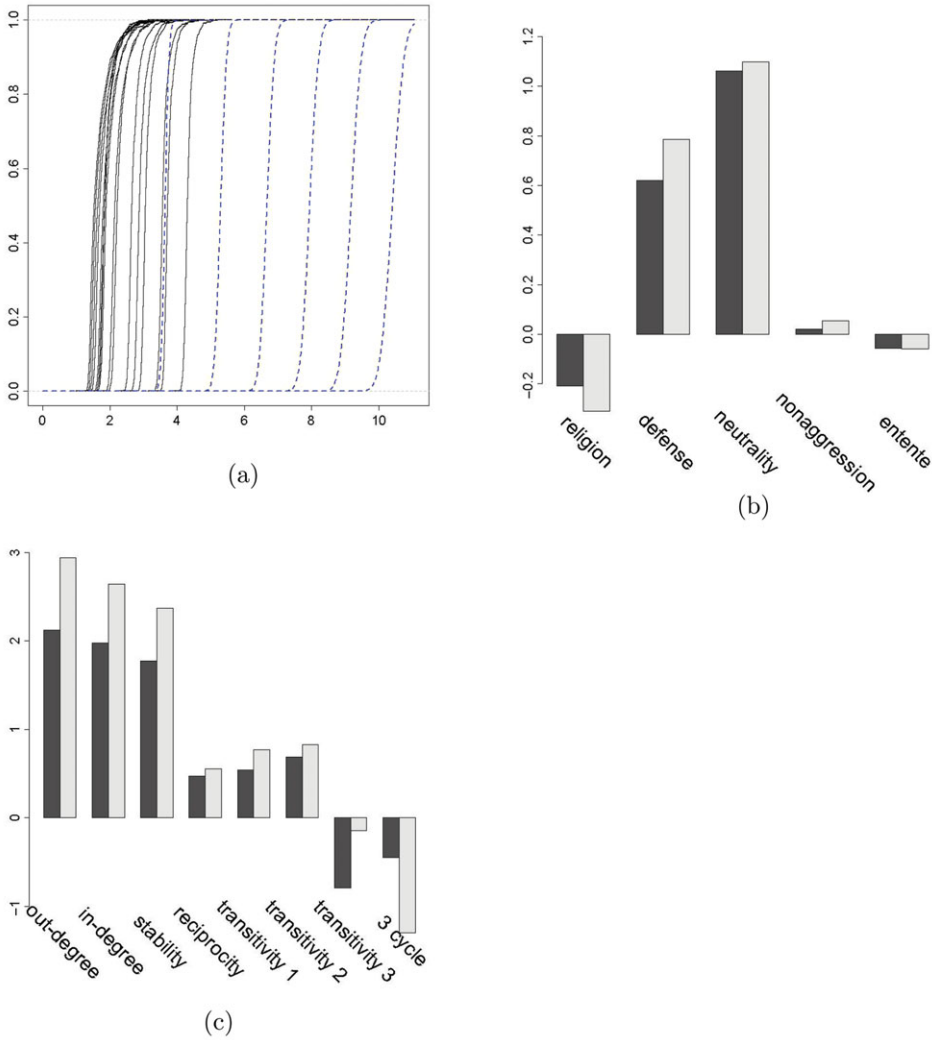


Fig. 5. Results from the world trade network data. (a) Plot of  $\mathcal{P}_{\epsilon,t}$  vs.  $\epsilon$ . Each curve corresponds to the random effects from a particular time point. The dotted lines correspond to the comparative random variable setting  $p = 0.05, 0.1, 0.15, 0.2, 0.25, 0.3$ . See Section 5 for details. (b) Posterior means of the covariates ( $\beta$ ). Dark gray indicates ignoring simultaneous dependence, while light gray indicates accounting for this dependence in the model. (c) Posterior means of the autoregressive terms ( $\theta$ ). Dark gray indicates ignoring simultaneous dependence, while light gray indicates accounting for this dependence in the model.

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7.2.2 Evaluating the effect of the time interval on simultaneous dependence

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As we have just seen, even at annual increments we see the presence of simultaneous dependence. We now show how this presence increases as the time intervals become coarser. We now consider the time interval from 1900 to 2000. This naturally diminishes the number of nations that exist during the entirety of the specified time

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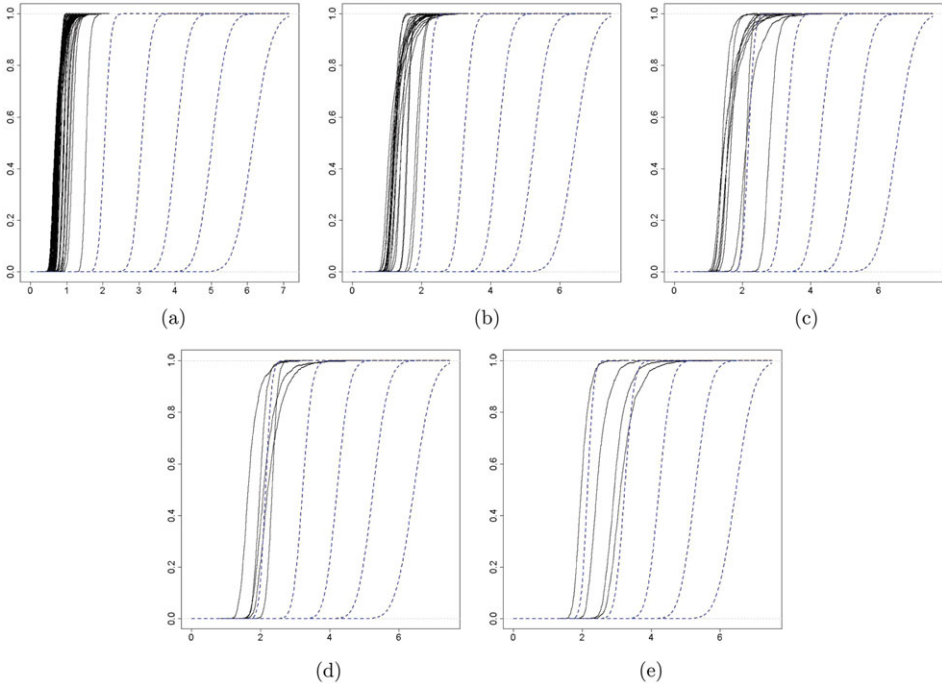


Fig. 6. World trade data: Plots of  $\mathcal{P}_{\epsilon,t}$  vs.  $\epsilon$ . Each curve corresponds to the random effects from a particular time point. The dotted lines in each figure correspond to the comparative random variable setting  $p = 0.05, 0.1, 0.15, 0.2, 0.25, 0.3$ . See Section 5 for details. Coarser time intervals lead to stronger evidence of simultaneous dependence. (a) Annual. (b) Every 5 years. (c) Every 10 years. (d) Every 20 years. (e) Every 25 years.

611 interval, and we are left with 28 nations. We apply our model to these 28 nations  
 612 looking at every year, every 5 years, every 10 years, every 20 years, and every 25  
 613 years. Intuition (as well as previous work by Lerner *et al.*, 2013) tells us that the  
 614 simultaneous dependence should grow as the time interval becomes larger, and in  
 615 fact this is what we see.

616 Figure 6 gives the evidence of the simultaneous dependence for the five data  
 617 sets. We can see that simultaneous dependence increases with the coarseness of the  
 618 time interval, as shown by the increasing trend for the location of the thin shell of  
 619 posterior probability mass for the individual effects. To corroborate this, we also  
 620 implemented the TERGM model on the five different data sets (collected every 1,  
 621 5, 10, 20, and 25 years). To capture the simultaneous dependencies, we included as  
 622 ERGM terms the counts of reciprocated ties, transitive triangles, and three-cycles.  
 623 Figure 7 shows the trends of these parameter estimates for the five data sets, where  
 624 the values for each parameter have been normalized by the corresponding parameter  
 625 value from the 25 year interval data. We see that the strength of the effect sizes  
 626 increase as the time between observations increases (we actually show the negative  
 627 of the three-cycle coefficients for visual clarity), thus corroborating our finding that  
 628 the simultaneous dependence does in fact increase.

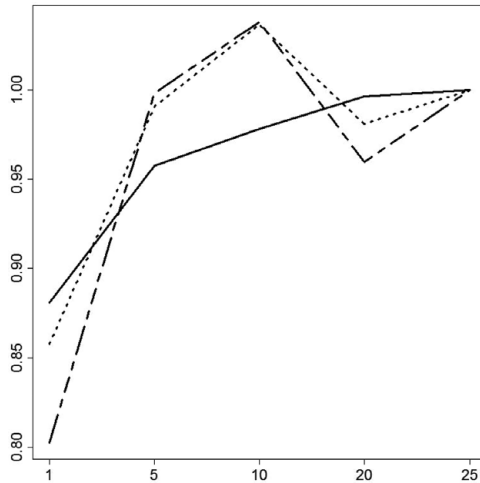


Fig. 7. TERGM coefficient estimates for reciprocity (solid), transitive triples (dotted), and cyclic triples (dash-dot) (negative coefficients given for the cyclic triples). Horizontal axis corresponds to the spacing of observations for the data set used. The increasing trend in the strength of the effect sizes corroborates our finding of increasing simultaneous dependence.

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## 8 Discussion

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In this paper, we have adapted the dynamic logistic network regression model of Almquist & Butts (2013) by introducing a framework for capturing not only temporal dependencies through an autoregressive mean structure but also simultaneous dependence through an autoregressive covariance structure. We demonstrated that ignoring simultaneous dependence leads to negative inferential consequences. The methods outlined here account for both complex temporal and simultaneous dependencies in a parsimonious way, while keeping within a familiar framework.

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Like many other statistical models for network data, scalability is an issue for all but very simple simultaneous dependence structures. While the VB estimation method proposed for the STAR model is quick for small to medium data sets, the requirement to invert large covariance matrices prohibits this methodology in its current state from being scaled up to extremely large networks.

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We have also described how our work may be placed within the familiar GLMM framework. While it is beyond the scope of this paper to thoroughly discuss model selection problems involving, e.g., covariance structures or link functions, it is the author's hope that previous and ongoing GLM and GLMM research (e.g., Chen & Tsurumi, 2010) can be used to build upon the proposed work in this area. Further, while we have shown practical operationalizations of the proposed method for binary data in Section 2.6, we leave it for future work to describe the specifics of sophisticated covariance structures (i.e.,  $H_{\cdot,t}$ 's that are more complicated than  $I_n$ ) for other data types.

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Other future work that would be valuable to the network analysis community would be to provide a thorough comparison of the available methods for discrete

654 temporal network data, such as the proposed approach, TERGM (Hanneke *et al.*,  
 655 2010) and STERGM (Krivitsky & Handcock, 2014), latent space models for  
 656 dynamic networks (Durante & Dunson, 2014; Sewell & Chen, 2015), and dynamic  
 657 stochastic blockmodels (Xing *et al.*, 2010). It would be important to know which  
 658 method ought to be used in various contexts, and under what circumstances the  
 659 conclusions from these models might differ.

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### Appendix A: Closed-form updates for VB

788 Before giving the closed form of the  $q$ 's, let us first provide a little notation that will  
 789 be used. Let  $I^- = J_n - I_n$ , i.e., the matrix of ones with zeros on the diagonal. Let  
 790  $tr(A)$  be the trace of some square matrix  $A$ . For a matrix  $\Sigma$ , let  $\Sigma_{(i,j)}$  denote the  $2 \times 2$   
 791 submatrix obtained from the  $i$ th and  $j$ th rows and columns. Let  $\mathcal{A}_t^-$  denote  $\text{vec}^-(A_t^*)$ .  
 792 Let  $trN(\boldsymbol{\mu}, \Sigma)$  be the truncated normal; we will not add any notation specifying the  
 793 varying domain as this should be obvious in our context from the data, which  $A_{ijt}^*$   
 794 are restricted to the positive reals and which to the negative reals. Finally, let  $\tilde{X}_t$   
 795 denote the  $n(n-1) \times (p_1 + p_2)$  matrix such that

$$\tilde{X}_t = (\text{vec}^-(X_{1t}), \dots, \text{vec}^-(X_{p_1t}), \text{vec}^-(\mathcal{G}_{1t}), \dots, \text{vec}^-(\mathcal{G}_{p_2t})).$$

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**Result 1.**  $q_1(\boldsymbol{\beta}, \boldsymbol{\theta}) \stackrel{\mathcal{Q}}{=} N(\boldsymbol{\mu}_m, \Sigma_m)$ , where

$$\Sigma_m^{-1} = \text{diag}(1/\sigma_{\beta}^2, \dots, 1/\sigma_{\beta}^2, 1/\sigma_{\theta}^2, \dots, 1/\sigma_{\theta}^2) + \sum_{t=1}^T \tilde{X}_t' \tilde{X}_t,$$

$$\boldsymbol{\mu}_m = \Sigma_m \left( \sum_{t=1}^T \tilde{X}_t' (M_{A_t} - \text{vec}^-(\boldsymbol{\mu}_{s_{1t}} + \boldsymbol{\mu}_{s_{2t}}) \mathbb{1}') - \text{vec}^-(\mathbb{1}(\boldsymbol{\mu}_{r_{1t}} + \boldsymbol{\mu}_{r_{2t}})') - \text{vec}^-(M_{R_t}) \right).$$

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**Result 2.**  $q_2(\tau_{s2}, \tau_{r2}, \Omega) \stackrel{\mathcal{D}}{=} IG(a_s, b_s)IG(a_r, b_r)IW(a_\Omega, B_\Omega)$  where

$$\begin{aligned} a_s &= a_{s0} + nT/2 & b_s &= b_{s0} + \frac{1}{2} \sum_{t=1}^T \left[ \text{tr}(\tilde{\Sigma}_{srt(s)} H_{st}^{-1}) + \boldsymbol{\mu}'_{s2t} H_{st}^{-1} \boldsymbol{\mu}_{s2t} \right] \\ a_r &= a_{r0} + nT/2 & b_r &= b_{r0} + \frac{1}{2} \sum_{t=1}^T \left[ \text{tr}(\tilde{\Sigma}_{srt(r)} H_{rt}^{-1}) + \boldsymbol{\mu}'_{r2t} H_{rt}^{-1} \boldsymbol{\mu}_{r2t} \right] \\ a_\Omega &= a_{\Omega 0} + nT & B_\Omega &= B_{\Omega 0} + \sum_{t=1}^n \sum_{i=1}^n \left[ \tilde{\Sigma}_{srt(sr)(i,n+i)} + (\boldsymbol{\mu}_{s1ti}, \boldsymbol{\mu}_{r1ti})' (\boldsymbol{\mu}_{s1ti}, \boldsymbol{\mu}_{r1ti}) \right], \end{aligned}$$

$\tilde{\Sigma}_{srt(s)}$  is the first  $n$  rows and first  $n$  columns of  $\tilde{\Sigma}_{srt}$ ,  $\tilde{\Sigma}_{srt(r)}$  is the second  $n$  rows and second  $n$  columns of  $\tilde{\Sigma}_{srt}$ , and  $\tilde{\Sigma}_{srt(sr)}$  is the last  $(2n)$  rows and  $(2n)$  columns of  $\tilde{\Sigma}_{srt}$ .

**Result 3.**  $q_3(\{\mathcal{A}_t\}_{t=1}^T) \stackrel{\mathcal{D}}{=} \prod_{t=1}^T \text{tr}N(M_{A_t}, I)$  where

$$M_{A_t} = \tilde{X}_t \boldsymbol{\mu}_m + \text{vec}^{-}((\boldsymbol{\mu}_{s1t} + \boldsymbol{\mu}_{s2t}) \mathbb{1}') + \text{vec}^{-}(\mathbb{1}(\boldsymbol{\mu}_{r1t} + \boldsymbol{\mu}_{r2t})') + \text{vec}^{-}(M_{R_t}).$$

**Result 4.**  $q_4(\{s_{1t}, r_{1t}, s_{2t}, r_{2t}\}_{t=1}^T) \stackrel{\mathcal{D}}{=} \prod_{t=1}^T N((\boldsymbol{\mu}'_{s1t}, \boldsymbol{\mu}'_{r1t}, \boldsymbol{\mu}'_{s2t}, \boldsymbol{\mu}'_{r2t})', \tilde{\Sigma}_{srt})$ , where

$$\begin{aligned} \tilde{\Sigma}_{srt}^{-1} &= \begin{pmatrix} 1 & 0 & 1 & 0 \\ 0 & 1 & 0 & 1 \\ 1 & 0 & 1 & 0 \\ 0 & 1 & 0 & 1 \end{pmatrix} \otimes (n-1)I_n + \begin{pmatrix} 0 & 1 & 0 & 1 \\ 1 & 0 & 1 & 0 \\ 0 & 1 & 0 & 1 \\ 1 & 0 & 1 & 0 \end{pmatrix} \otimes I^- \\ &+ \left( \begin{array}{c|c} a_\Omega B_\Omega^{-1} \otimes I_n & 0 \\ \hline 0 & \begin{matrix} \frac{a_s}{b_s} H_{s1t}^{-1} & 0 \\ 0 & \frac{a_r}{b_r} H_{r1t}^{-1} \end{matrix} \end{array} \right) \end{aligned}$$

$$\begin{pmatrix} \boldsymbol{\mu}_{s1t} \\ \boldsymbol{\mu}_{r1t} \\ \boldsymbol{\mu}_{s2t} \\ \boldsymbol{\mu}_{r2t} \end{pmatrix} = \tilde{\Sigma}_{srt} \begin{pmatrix} \left( \text{rev-vec}^{-}(M_{A_t} - \tilde{X}_t \boldsymbol{\mu}_m) - M_{R_t} \right) \mathbb{1} \\ \left( \text{rev-vec}^{-}(M_{A_t} - \tilde{X}_t \boldsymbol{\mu}_m)' - M_{R_t} \right) \mathbb{1} \\ \left( \text{rev-vec}^{-}(M_{A_t} - \tilde{X}_t \boldsymbol{\mu}_m) - M_{R_t} \right) \mathbb{1} \\ \left( \text{rev-vec}^{-}(M_{A_t} - \tilde{X}_t \boldsymbol{\mu}_m)' - M_{R_t} \right) \mathbb{1} \end{pmatrix}$$

and  $\text{rev-vec}^{-}(\cdot)$  is the matrix (with zero diagonal elements) constructed by reversing the  $\text{vec}^{-}(\cdot)$  operator.

*Derivation:*

We first provide some preliminary results:

1. For some  $n \times 1$  vectors  $\mathbf{a}_1$  and  $\mathbf{a}_2$ ,  $\text{tr}(D_{a_1} I^{-1} D_{a_2}) = (n-1) \mathbf{a}'_1 \mathbf{a}_2$ , where  $D_a$  denotes a diagonal matrix whose entries are  $\mathbf{a}$ .
2. For some  $n \times n$  matrix  $A$ ,  $\text{tr}(I^{-1} D_a (A \circ I^{-1})) = \mathbf{a}' (A \circ I^{-1}) \mathbb{1}$ .
3.  $\text{tr}(I^{-1} D_{a_1} I^{-1} D_{a_2}) = \mathbf{a}'_1 I^{-1} \mathbf{a}_2$ .

Also note that since  $\text{vec}^{-}(A)' \text{vec}^{-}(A) = \text{vec}(A \circ I^{-1})' \text{vec}(A \circ I^{-1}) = \text{tr}((A \circ I^{-1})'(A \circ I^{-1}))$ , we may consider the conditional probability of  $\mathcal{A}_t | s_{1t}, r_{1t}, s_{2t}, r_{2t}, \cdot$  as proportional (with respect to the sender and receiver effects) to the matrix normal distribution kernel of  $A_t^* \circ I^{-1}$ .

817 Letting  $\tilde{A}_t = (A_t^* - \langle \boldsymbol{\beta}, \mathcal{X}_t \rangle + \langle \boldsymbol{\theta}, \mathcal{G}_t \rangle) \circ I^-$ , we have, dropping the subscript  $t$ ,

$$\begin{aligned} & \log(\pi(A^* | s_1, \mathbf{r}_1, s_2, \mathbf{r}_2, \cdot)) \\ &= \text{const} - \frac{1}{2} \text{tr} [(\tilde{A} - D_{s_1} I^- - D_{s_2} I^- - I^- D_{r_1} - I^- D_{r_2})' \\ & \quad \times (\tilde{A} - D_{s_1} I^- - D_{s_2} I^- - I^- D_{r_1} - I^- D_{r_2})] \\ &= \text{const} - \frac{1}{2} \text{tr} [I^- D_{s_1} D_{s_1} I^- - 2I^- D_{s_1} \tilde{A} + 2I^- D_{s_1} D_{s_2} I^- + 2I^- D_{s_1} I^- D_{r_1} \\ & \quad + 2I^- D_{s_1} I^- D_{r_2} - 2I^- D_{s_2} \tilde{A} + I^- D_{s_2} D_{s_2} I^- + 2I^- D_{s_2} I^- D_{r_1} + 2I^- D_{s_2} I^- D_{r_2} \\ & \quad + D_{r_1} I^- I^- D_{r_1} + 2D_{r_1} I^- I^- D_{r_2} + D_{r_2} I^- I^- D_{r_2} - 2D_{r_1} I^- \tilde{A} - 2D_{r_2} I^- \tilde{A}] \\ &= \text{const} - \frac{1}{2} [(n-1)s'_1 s_1 - 2s_1 \tilde{A} \mathbb{1} + 2(n-1)s'_1 s_2 + 2s'_1 I^- \mathbf{r}_1 + 2s'_1 I^- \mathbf{r}_2 - 2s'_2 \tilde{A} \mathbb{1} \\ & \quad + (n-1)s'_2 s_2 + 2s'_2 I^- \mathbf{r}_1 + 2s'_2 I^- \mathbf{r}_2 + (n-1)r'_1 \mathbf{r}_1 + 2(n-1)r'_1 \mathbf{r}_2 + (n-1)r'_2 \mathbf{r}_2 \\ & \quad - 2r'_1 \tilde{A}' \mathbb{1} - 2r'_2 \tilde{A}' \mathbb{1}]. \end{aligned}$$

818 Combining the expected value of this under  $q$  with  $\mathbb{E}_q(\log(\pi(s_{1t}, \mathbf{r}_{1t}, s_{2t}, \mathbf{r}_{2t} | \tau_{s_2}, \tau_{r_2}, \boldsymbol{\Omega},$   
819  $A_{t-1})))$  yields Result 4.  $\square$

820 **Result 5.**  $q_5(\{\mathcal{R}_t\}_{t=1}^T) \stackrel{\mathcal{D}}{=} \prod_t \prod_{i < j} N(M_{R_t}[i, j], \tilde{\sigma}_R^2)$  where

$$\begin{aligned} M_{R_t}[i, j] &= \tilde{\sigma}_R^2 (\tilde{A}_{ijt} + \tilde{A}_{jit}), \\ \tilde{\sigma}_R^2 &= \frac{b_R / a_R}{1 + 2b_R / a_R}, \\ \tilde{A}_{ijt} &= \text{rev-vec} \left( M_{A_t} - \tilde{X}_t \boldsymbol{\mu}_m \right) [i, j] - \boldsymbol{\mu}_{s_{1t}}[i] - \boldsymbol{\mu}_{s_{2t}}[i] - \boldsymbol{\mu}_{r_{1t}}[j] - \boldsymbol{\mu}_{r_{2t}}[j]. \end{aligned}$$

821 For the purposes of computing the parameters for the other  $q$ 's, assume for  $i < j$  that  
822  $M_{R_t}[j, i] = M_{R_t}[i, j]$ .

823 **Result 6.**  $q_6(\sigma_R^2) \stackrel{\mathcal{D}}{=} IG(a_R, b_R)$  where

$$\begin{aligned} a_R &= a_{R0} + \frac{Tn(n-1)}{4} \\ b_R &= b_{R0} + \frac{1}{2} \sum_t \sum_{i < j} (\tilde{\sigma}_R^2 + M_{R_t}[i, j]^2) \end{aligned}$$

824 **Result 7.** For the undirected case,  $q_4(\{\mathbf{s}_t\}_{t=1}^T) = \prod_{t=1}^T N(\boldsymbol{\mu}'_{st}, \tilde{\Sigma}_{st})$ , where

$$\begin{aligned} \boldsymbol{\mu}_{st} &= \tilde{\Sigma}_{st} \mathbf{E}(A_t^* \circ I^-) \mathbb{1} \\ \tilde{\Sigma}_{st}^{-1} &= (n-1)I_n + I^- + \frac{a_s}{b_s} H_{st}^{-1} \end{aligned}$$

825 *Derivation:* Define  $I^\Delta$  as the square matrix with ones on the upper triangle and zero  
826 everywhere else (the diagonal is also zero). As before, it is helpful to provide some  
827 preliminary results:

- 828 1. For some  $n \times 1$  vector  $\mathbf{a}$ ,  $\text{tr}(D_{\mathbf{a}}(I^\Delta I^{\Delta'} + I^{\Delta'} I^\Delta) \mathbf{a}) = (n-1) \mathbf{a}' \mathbf{a}$ .  
829 2. For some  $n \times n$  matrix  $A$ ,

$$\text{tr}(D_{\mathbf{a}}(\tilde{A}' I^\Delta + \tilde{A} I^{\Delta'})) = \text{tr}(D_{\mathbf{a}} I^- A) = \mathbf{a}' (A \circ I^-) \mathbb{1}.$$

- 830 3.  $2 \cdot \text{tr}(D_{\mathbf{a}} I^{\Delta'} D_{\mathbf{a}} I^\Delta) = \mathbf{a}' I^- \mathbf{a}$ .

To show this last, note that the  $i$ th diagonal of  $D_{\mathbf{a}}I^{\Delta'}D_{\mathbf{a}}I^{\Delta} = \sum_{j=1}^{i-1} \mathbf{a}_i\mathbf{a}_j$ , and hence the trace equals  $\sum_{i=1}^n \sum_{j=1}^{i-1} \mathbf{a}_i\mathbf{a}_j = \mathbf{a}'I^{\Delta'}\mathbf{a} = \mathbf{a}'I^{\Delta}\mathbf{a}$ . This then implies that  $2 \cdot \text{tr}(D_{\mathbf{a}}I^{\Delta'}D_{\mathbf{a}}I^{\Delta}) = \mathbf{a}'I^{\Delta'}\mathbf{a} + \mathbf{a}'I^{\Delta}\mathbf{a} = \mathbf{a}'I^{-}\mathbf{a}$ .

Let  $\tilde{A}_t = (A_t^* - \langle \boldsymbol{\beta}, \mathcal{X}_t \rangle - \langle \boldsymbol{\theta}, \mathcal{G}_t \rangle) \circ I^{\Delta}$ . Then we have, dropping the subscript  $t$ ,

$$\begin{aligned} \log(\pi(A_t^*|\mathbf{s})) &= \text{const} - \frac{1}{2} \text{tr} [(\tilde{A} - D_s I^{\Delta} - I^{\Delta} D_s)'(\tilde{A} - D_s I^{\Delta} - I^{\Delta} D_s)] \\ &= \text{const} - \frac{1}{2} \text{tr} [D_s(I^{\Delta}I^{\Delta'} + I^{\Delta'}I^{\Delta})D_s + 2D_sI^{\Delta'}D_sI^{\Delta} - 2D_s(\tilde{A}'I^{\Delta} + \tilde{A}I^{\Delta'})] \\ &= \text{const} - \frac{1}{2} \left[ \mathbf{s}' \left( (n-1)I + I^{-} + \frac{1}{\tau_s} H_s^{-1} \right) \mathbf{s} - 2\mathbf{s}'(A^* \circ I^{-})\mathbf{1} \right]. \end{aligned}$$

Combining the expected value of this under  $q$  with  $\mathbb{E}_q(\log(\pi(\mathbf{s}_t|\tau_s, A_{t-1})))$  yields Result 7.  $\square$

## Appendix B: Proofs

### B.1 Proposition of Section 2.3

*Proof.* Letting  $m_{ijt} = \langle \boldsymbol{\beta}, \mathcal{X}_t \rangle[i, j] + \langle \boldsymbol{\theta}, \mathcal{G}_t \rangle[i, j]$  and  $V := \text{Var}(s_{it} + r_{jt})$ , we have

$$\begin{aligned} \mathbb{P}(A_{ijt} = 1 | \boldsymbol{\beta}, \boldsymbol{\theta}) &= \mathbb{E} \left( \mathbb{E}(A_{ijt} | s_{it} + r_{jt}, \boldsymbol{\beta}, \boldsymbol{\theta}) | \boldsymbol{\beta}, \boldsymbol{\theta} \right) \\ &= \mathbb{E} \left( \Phi \left( \frac{s_{it} + r_{jt} + m_{ijt}}{\sqrt{\text{Var}(E_{ijt})}} \right) | \boldsymbol{\beta}, \boldsymbol{\theta} \right) \\ &= \int_{-\infty}^{\infty} \int_{-\infty}^{\frac{s_{it} + r_{jt} + m_{ijt}}{\sqrt{\text{Var}(E_{ijt})}}} \frac{1}{\sqrt{2\pi}} e^{-\frac{z^2}{2}} \frac{1}{\sqrt{2\pi V}} e^{-\frac{(s_{it} + r_{jt})^2}{2V}} dZ d(s_{it} + r_{jt}) \\ &= \mathbb{P}(Z \sqrt{\text{Var}(E_{ijt})} - (s_{it} + r_{jt}) < m_{ijt}). \end{aligned}$$

Since  $Z \sqrt{\text{Var}(E_{ijt})} - (s_{it} + r_{jt}) \sim N(0, \text{Var}(E_{ijt}) + V)$ , our result holds.  $\square$

### B.2 Theorem of Section 2.4

*Proof.* It is obvious that the mean of each  $A_{ijt}^*$  are equivalent for (I), (II), and (III), and that the covariance between any  $A_{ijt}^*$  and  $A_{k'lt}^*$  as given by (III) satisfies Equation (4).

It is straightforward to check that  $\sigma_R^2 M_R + (\sigma_{\epsilon}^2 + \sigma_R^2) I_{n^2}$  satisfies the final two terms in Equation (4), and that this is the covariance matrix of  $\text{vec}(E_t)$ . Note that for any two  $n$ -dimensional vectors  $\mathbf{a}$  and  $\mathbf{b}$ , we have that

1.  $\text{vec}(\mathbf{a}\mathbf{b}') = \mathbf{b} \otimes \mathbf{a}$ ,
2.  $\text{Cov}(\mathbf{1} \otimes \mathbf{a}) = J_n \otimes \text{Cov}(\mathbf{a})$ ,
3.  $\text{Cov}(\mathbf{a} \otimes \mathbf{1}) = \text{Cov}(\mathbf{a}) \otimes J_n$ , and
4.  $\text{Cov}(\mathbf{1} \otimes \mathbf{a}, \mathbf{b} \otimes \mathbf{1}) = \mathbf{1} \otimes \text{Cov}(\mathbf{a}, \mathbf{b}) \otimes \mathbf{1}'$ ,

852 where  $J_n$  is the  $n \times n$  matrix of 1's. We may then write the covariance of the  $A_{ijt}^*$ 's  
853 as given in (III) as

$$\begin{aligned} \text{Cov}(\mathcal{A}_t) &= \text{Cov}(\text{vec}(s_t \mathbb{1}') + \text{vec}(\mathbb{1} r_t') + \text{vec}(E_t)) \\ &= \text{Cov}(\mathbb{1} \otimes s_t + r_t \otimes \mathbb{1} + \text{vec}(E_t)) \\ &= J_n \otimes \Sigma_{st} + \Sigma_{rt} \otimes J_n + \mathbb{1} \otimes \Sigma_{srt} \otimes \mathbb{1}' + \mathbb{1}' \otimes \Sigma'_{srt} \otimes \mathbb{1} + \sigma_R^2 M_R + (\sigma_\epsilon^2 + \sigma_R^2) I_{n^2}. \end{aligned}$$

854 Hence (I), (II), and (III) have the same covariance structure.

855 Finally, we have from (III)

$$\begin{aligned} \mathcal{A}_t &= \text{vec}(\langle \boldsymbol{\beta}, \mathcal{X}_t \rangle + \langle \boldsymbol{\theta}, \mathcal{G}_t \rangle) + \mathbb{1} \otimes s_t + r_t \otimes \mathbb{1} + \text{vec}(E_t) \\ &= \text{vec}(\langle \boldsymbol{\beta}, \mathcal{X}_t \rangle + \langle \boldsymbol{\theta}, \mathcal{G}_t \rangle) + (\mathbb{1} \otimes I_n) s_t + (I_n \otimes \mathbb{1}) r_t + \text{vec}(E_t) \\ &\stackrel{\mathcal{D}}{=} \text{vec}(\langle \boldsymbol{\beta}, \mathcal{X}_t \rangle + \langle \boldsymbol{\theta}, \mathcal{G}_t \rangle) + \left( (\mathbb{1} \otimes I_n, I_n \otimes \mathbb{1}) \Sigma_t^{\frac{1}{2}}, (\sigma_R^2 M_R + (\sigma_\epsilon^2 + \sigma_R^2) I_{n^2})^{\frac{1}{2}} \right) \mathbf{z} \end{aligned}$$

856 where  $\mathbf{z}$  is a  $(2n + n^2) \times 1$  vector of independent standard normal random variables,  
857 and

$$\Sigma_t := \begin{pmatrix} \Sigma_{st} & \Sigma_{srt} \\ \Sigma'_{srt} & \Sigma_{rt} \end{pmatrix},$$

858 Since  $\text{vec}(\mathcal{A}_t)$  is an affine transformation of  $\mathbf{z}$ , we have that the  $A_{ijt}^*$ 's are jointly  
859 normal, indicating that (I), (II), and (III) are equivalent.  $\square$